# Final Report – Melt Rate Enhancement for High Aluminum HLW Glass Formulations, VSL-08R1360-1, Rev. 0, dated 12/19/08

Prepared for the U.S. Department of Energy Assistant Secretary for Environmental Management



P.O. Box 450 Richland, Washington 99352

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### **Final Report**

# Melt Rate Enhancement for High Aluminum HLW Glass Formulations

prepared by

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This report describes the results of testing specified by the above Test Plan The work was performed in compliance with the quality assurance requirements specified in the Test Plan. Results required by the Test Plan are reported. The test results and this report have been reviewed for correctness, technical adequacy, completeness, and accuracy.

I.L. Pegg: VSL Program Director/Principal Investigator

Date: 12/19/08

I. Joseph: EnergySolutions Sub-Contract Manager

08 Date:

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#### List of Abbreviations

AA	Atomic Absorption Spectroscopy
ADS	Air Displacement Slurry
ADS ANL-LRM	Argonne National Laboratory – Low Activity Waste Reference Material
AOD	Air Operated Diaphragm
CCC	Canister Center Line Cooling
DCP-AES	
DCP-AES DF	Direct Current Plasma Atomic Emission Spectroscopy Decontamination Factor
DF DFC	
	DM10 Feed Consumption (test) DuraMelter®
DM	
DOE DRE	Department of Energy
	Destruction & Removal Efficiency
DWPF	Defense Waste Processing Facility
EA	Environmental Assessment
EPA	Environmental Protection Agency
EDS	Energy Dispersive X-Ray Spectroscopy
FTIR	Fourier Transform Infrared Spectroscopy
GC	Gas Chromatography
HEPA	High-Efficiency Particulate Air Filter
HEME	High-Efficiency Mist Eliminator
HLW	High Level Waste
IHLW	Immobilized High Level Waste
LAW	Low Activity Waste
M	Molarity
N	Normality
NIST	National Institute of Standards and Technology
ORP	Office of River Protection
PBS	Packed Bed Scrubber
PCT	Product Consistency Test
QA	Quality Assurance
QAPjP	Quality Assurance Project Plan for Testing Programs Generating Environmental Regulatory Data
QAPP	Quality Assurance Project Plan
RPP	River Protection Project
SBS	Submerged Bed Scrubber
SEM	Scanning Electron Microscopy
SIPP	Semi Integrated Pilot Plant
SOP	Standard Operating Procedure
SRM	Standard Reference Material
SRS	Savannah River Site
TCLP	Toxicity Characteristic Leaching Procedure
TCO	Thermal Catalytic Oxidation
VGF	Vertical Gradient Furnace
VSL	Vitreous State Laboratory
W.C.	Water Column
WESP	Wet Electrostatic Precipitator
WTP	Hanford Tank Waste Treatment and Immobilization Plant
WVDP	West Valley Demonstration Project
XRD	X-ray Diffraction
XRF	X-Ray Fluorescence Spectroscopy

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#### SECTION 1.0 INTRODUCTION

This report describes the development and testing of new glass formulations for high aluminum waste streams that achieve high waste loadings while maintaining high processing rates. The testing was based on the compositions of Hanford High Level Waste (HLW) with limiting concentrations of aluminum specified by the Office of River Protection (ORP). The testing identified glass formulations that optimize waste loading and waste processing rate while meeting all processing and product quality requirements. The work included preparation and characterization of crucible melts and small scale melt rate screening tests. The results were used to select compositions for subsequent testing in a DuraMelter 100 (DM100) system. These tests were used to determine processing rates for the selected formulations as well as to examine the effects of increased glass processing temperature, and the form of aluminum in the waste simulant. Finally, one of the formulations was selected for large-scale confirmatory testing on the HLW Pilot Melter (DM1200), which is a one third scale prototype of the Hanford Tank Waste Treatment and Immobilization Plant (WTP) HLW melter and off-gas treatment system. This work builds on previous work performed at the Vitreous State Laboratory (VSL) for Department of Energy (DOE) to increase waste loading and processing rates for high-iron HLW waste streams [1] as well as previous tests conducted for ORP on the same high-aluminum waste composition used in the present work and other Hanford HLW compositions [2]. The scope of this study was outlined in a Test Plan [3] that was prepared in response to an ORP-supplied statement of work [4].

It is currently estimated that the number of HLW canisters to be produced in the WTP is about 13,500 (equivalent to 40,500 MT glass) [4]. This estimate is based upon the inventory of the tank wastes, the anticipated performance of the sludge treatment processes, and current understanding of the capability of the borosilicate glass waste form. The WTP HLW melter design, unlike earlier DOE melter designs, incorporates an active glass bubbler system. The bubblers create active glass pool convection and thereby improve heat transfer and glass melting rate. The WTP HLW melter has a glass surface area of  $3.75 \text{ m}^2$  and depth of  $\sim 1.1 \text{ m}$ . The two melters in the HLW facility together are designed to produce up to 7.5 MT of glass per day at 100% availability. Further increases in HLW waste processing rates can potentially be achieved by increasing the melter operating temperature above  $1150^{\circ}$ C and by increasing the waste loading in the glass product. Increasing the waste loading also has the added benefit of decreasing the number of canisters for storage.

The current estimates and glass formulation efforts have been conservative in terms of achievable waste loadings. These formulations have been specified to ensure that the glasses are homogenous, contain essentially no crystalline phases, are processable in joule-heated, ceramic-lined melters and meet WTP Contract terms. The WTP's overall mission will require the immobilization of tank waste compositions that are dominated by mixtures of aluminum (Al), chromium (Cr), bismuth (Bi), iron (Fe), phosphorous (P), zirconium (Zr), and sulfur (S) compounds as waste-limiting components. Glass compositions for these waste mixtures have been developed based upon previous experience and current glass property models. Recently,

DOE has initiated a testing program to develop and characterize HLW glasses with higher waste loadings [4, 5]. Results of this work have demonstrated the feasibility of increases in waste-loading from about 25 wt% to 33-50 wt% (based on oxide loading) in the glass depending on the waste stream. It is expected that these higher waste loading glasses will reduce the HLW canister production requirement by about 25% or more [2].

DOE HLW treatment programs have featured joule heated ceramic melter technology for the vitrification of high level tank waste. The melter technology used at the West Valley Demonstration Project (WVDP) in New York and at the Defense Waste Processing Facility (DWPF) at the Savannah River Site (SRS) process(ed) HLW in ceramic melters at an operating temperature of 1150°C. Historically, HLW melters are operated at temperatures of 1150°C to allow for sufficient temperature control for normal as well as upset conditions in an operating melter, while still protecting the electrodes from potential damage due to unanticipated high temperature swings. Since the HLW melters deployed in the United States at West Valley and DWPF do not actively mix the glass pool, temperature variations within the glass pool can be relatively large (~  $\pm 75^{\circ}$ C) with respect to the nominal operating temperature since natural convection within the glass pool is limited in the viscous molten glass. In advancing the technology, EnergySolutions/VSL have demonstrated on very large scale melters (Energy Solutions M-Area facility, RPP-WTP HLW Pilot Melter, and the RPP-WTP LAW Pilot Melter) that active mixing of the glass pool using our patented bubbler technology significantly reduces the temperature gradient within the glass pool and allows the melter to be controlled in a tighter operating band. As a result, the operating temperature of the melter can be modestly increased to about 1175°C with the current materials of construction, (and up to 1225°C with changes of electrode and bubbler materials) while maintaining the operating integrity of the melter at the higher temperature. Tests conducted with various HLW waste streams on the DM100 and DM1200 melters have demonstrated increases in glass production rates from 0 to 225 percent while increasing the processing temperature from 1150°C to 1175°C [2, 6]. Further increases in operating temperature to higher temperatures (1200°C) have the potential to further increase processing rate as well as increased waste loading, both of which translate into significant cost savings.

Under a separate contract to support the WTP, the VSL is developing and testing glass formulations for WTP HLW waste compositions to provide data to meet the WTP contract requirements and to support system design activities [7-13]. That work is based upon small-scale batch melts ("crucible melts") using waste simulants. Selected formulations have also been tested in small-scale, continuously fed, joule-heated melters (DM100) [14-17] and, ultimately, in the HLW DM1200 Pilot Melter [6, 16-23]. More recently, a series of tests were conducted on the DM100 to determine the effect of systematic variations in various glass properties (viscosity and conductivity) and oxide concentrations on glass production rates with HLW feed streams [24, 25]. Such melter tests provide information on key process factors such as feed processing behavior, dynamic effects during processing, secondary phase formation, processing rates, off-gas amounts and compositions, foaming control, etc., that cannot be reliably obtained from crucible melts. This sequential scale-up approach in the vitrification testing program ensures that maximum benefit is obtained from the more costly larger-scale melter tests and that the most effective use is made of those test resources.

The glass formulation and melter testing work described in this report led to the identification of glass compositions that achieve high processing rates while maintaining high waste loadings of a high-aluminum HLW stream; that HLW composition was one of the four waste streams previously specified by ORP [5]. The previous tests with these waste streams demonstrated substantial increases in waste loading; however, production rates with the aluminum limited waste stream and with a waste stream limited by aluminum in combination with sodium were only a third to a half of the rates obtained for chromium, and iron limited wastes. The goal of the current work was, therefore, to retain high waste loadings and acceptable glass properties for the aluminum limited waste while increasing the glass production rate through the manipulation of the glass formulations and glass forming additives. An extensive literature review was performed during the first phase of this work, the results of which were included in the final report [2]. In the present work, the literature was further reviewed with particular emphasis on factors affecting melt rates. Although directly relevant literature on increasing the processing rates of high aluminum waste vitrification feeds is very limited, the more general literature on melt rates and feed to glass conversion processes (see for example [26 - 43]) provided a useful starting point for the present work.

The results from this test program will provide ORP with a basis for projection of the amount of Immobilized High Level Waste (IHLW) to be produced at Hanford and the potential waste processing rate, and evaluation of the likely potential for future enhancements of the WTP over and above the present well-developed baseline. It should be noted that the compositions of the four ORP-specified waste streams differ significantly from those of the feed tanks (AZ-101, AZ-102, C-16/AY-102, and C-104/AY-101) that have been the focus of the extensive technology development and design work performed for the WTP baseline. In this regard, the work detailed in this report is complementary to and necessarily of a more exploratory nature than the work in support of the current WTP baseline. It should be noted, therefore, that considerable further work would be required to bring the level of confidence in the new glass composition regions to a similar level of maturity to that of the current WTP baseline. Additional testing at the crucible and melter scales will be needed to determine the robustness of the new compositions with respect to variations in the feed compositions that may results from process variations as well as for the collection of data to support engineering and permitting requirements using a WTP prototypic off-gas system. In addition, since the high waste loading glass compositions are in a new composition space as compared to the current WTP compositions, additional effort will be required to develop and extend the current qualified glass composition region and supporting models to include these new compositions.

#### 1.1 Test Objectives

The principal objective of the work reported here was to develop and identify HLW glass compositions that maximize waste processing rates for the aluminum limted waste composition specified by ORP while maintaining high waste loadings and acceptable glass properties. This was accomplished through a combination of crucible-scale tests, confirmation tests on the DM100 melter system, and demonstration at pilot scale (DM1200). The DM100-BL unit was selected for these tests since it was used previously with the HLW waste streams evaluated in this study [2], was used for tests on HLW glass compositions [14-17] to support subsequent tests

on the HLW Pilot Melter [6, 16-23], conduct tests to determine the effect of various glass properties (viscosity and conductivity) and oxide concentrations on glass production rates with HLW feed streams [24, 25], and to assess the volatility of cesium and technetium during the vitrification of an HLW AZ-102 composition [44]. The same melter was selected for the present tests in order to maintain comparisons between the previously collected data. These tests provide information on melter processing characteristics and off-gas data, including formation of secondary phases and partitioning. Once DM100 tests were completed, one of the compositions was selected for further testing on the DM1200; the DM1200 system has been used for processing a variety of simulated Hanford waste streams [6, 16-23]. Tests on the larger melter provide processing data at one third of the scale of the actual WTP HLW melter and, therefore, provide a more accurate and reliable assessment of production rates and potential processing issues.

The work focused on maximizing waste processing rates for high aluminum HLW compositions. In view of the diversity of forms of aluminum in the Hanford tanks, tests were also conducted on the DM100 to determine the effect of changes in the form of aluminum on feed properties and production rate. In addition, the work evaluated the effect on production rate of modest increases in melter operating temperature.

Glass composition development was based on one of the HLW waste compositions specified by ORP that has a high concentration of aluminum. Small-scale tests were used to provide an initial screening of various glass formulations with respect to melt rates; more definitive screening was provided by the subsequent DM100 tests. Glass properties evaluated included: viscosity, electrical conductivity, crystallinity, gross glass phase separation and the 7-day Product Consistency Test (ASTM-1285). Glass property limits were based upon the reference properties for the WTP HLW melter. However, the WTP crystallinity limit (< 1 vol% at 950°C) was relaxed slightly as a waste loading constraint for the crucible melts.

#### **1.2 Quality Assurance**

This work was conducted under a quality assurance program that is based on NQA-1 (1989) and NQA-2a (1990) Part 2.7 that is in place at the VSL. Recently, the quality assurance program was updated to be compliant with applicable criteria of 10 CFR 830.120; Office of Civilian Waste Management DOE/RW-0333P, Quality Assurance Requirements and Description (QARD) Revision 18; the American Society of Mechanical Engineers (ASME) NQA-1, 2000; and DOE Order 414.1 C, Quality Assurance. This program is supplemented by a Quality Assurance Project Plan for RPP-WTP work [45] that is conducted at VSL. Test and procedure requirements by which the testing activities are planned and controlled are also defined in this plan. The program is supported by VSL standard operating procedures that were used for this work [46]. In addition, the requirements of DOE/RW-0333P were applicable to the following specific aspects of this work:

- Crucible melt preparation
- Analysis of crucible melt glasses
- **PCT**

#### 1.3 DM100 Melter System

#### 1.3.1 DM100 Feed System

A schematic diagram of the DM100 vitrification system is shown in Figure 1.1. The melter feed is introduced in batches into a feed container that is mounted on a load cell for weight monitoring. The feed is stirred with a variable speed mixer and constantly recirculated except for periodic, momentary interruptions during which the weight is recorded. Feed is normally introduced into the melter via a system designed to mimic the operation of an Air Displacement Slurry (ADS) pump, which is the present WTP baseline; however, a peristaltic pump was used in these tests in order to provide a uniform delivery of feed to the melt surface. In this system, a recirculation loop extends to the top of the melter where feed is diverted from the recirculation loop to the peristaltic pump and subsequently into the melter through a Teflon-lined feed line and water-cooled, vertical feed tube.

#### 1.3.2 Melter System

Cross-sectional diagrams of the DM100-BL melter are shown in Figures 1.2.a-c. The DM100-BL unit is a ceramic refractory-lined melter fitted with five electrodes: two pairs of opposing Inconel 690 plate electrodes and a bottom electrode. Power can be supplied in either three-phase or single-phase configurations. All of the tests in the present work were performed with the upper and lower electrodes on each side connected together and powered by a single-phase supply; the bottom electrode was not powered. Melt pool agitation is achieved by either a removable lance entering from the top of the melter or a permanent bubbler installed through the bottom electrode. In these tests the lance bubbler was used. The glass product is removed from the melter by means of an airlift discharge system. The melter has a melt surface area of 0.108 m<sup>2</sup> and a variable glass inventory of between 110 kg, when only the bottom pair of electrodes is used, and about 170 kg when both pairs of electrodes are used, which was the case in the present tests.

#### 1.3.3 Off-Gas System

For operational simplicity, the DM100-BL is equipped with a dry off-gas treatment system involving gas filtration operations only. Exhaust gases leave the melter plenum through a film cooler device that minimizes the formation of solid deposits. The film-cooler air has constant flow rate and its temperature is thermostatically controlled. Consequently, under steady-state operating conditions, the exhaust gases passing through the transition line (between the melter and the first filtration device) can be sampled at constant temperature and airflow rate. The geometry of the transition line conforms to the requirements of the 40-CFR-60 air sampling techniques. Immediately downstream of the transition line are cyclonic filters followed by conventional pre-filters and HEPA filters. The temperature of the cyclonic filters is maintained above 150°C while the temperatures in the HEPAs are kept sufficiently high to prevent moisture condensation. The entire train of gas filtration operations is duplicated and each train is used alternately. An induced draft fan completes the system.

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#### 1.4 DM1200

#### 1.4.1 Feed System

The feed material for these tests was prepared and controlled according to VSL specifications by a chemical supplier, as detailed in Section 2. Each batch of feed slurry was shipped to VSL in lined 55-gallon drums, which were staged for unloading into the mix tank. Both the mix tank and the feed tank are 750-gallon polyethylene tanks with conical bottoms that are fitted with mechanical agitators; the feed tank is also fitted with baffles to improve mixing. Any required feed additive can be added to the mix tank. Five calibrated load cells directly mounted on the legs of the feed tank are used to measure additions to, and removal from, the feed tank and are electronically monitored to determine the feed rate to the melter. The requisite amount of feed is pumped to the feed tank from the mix tank; measured amounts of water are combined by weight with the feed at this point to adjust the concentration of the melter feed. The material in the feed tank is constantly recirculated from the feed tank discharge outlet, at the tank bottom, to the tank inlet at the top, which provides additional mixing.

The feed is introduced into the melter using an ADS pump, which is the present WTP baseline. The feed transfer line extends from the outlet of the ADS pump in the feed tank to the top of the melter. Feed is introduced into the melter through an un-cooled feed nozzle that is located above the center of the glass pool. Only one feed tube is used to represent the planned number of feed tubes per unit melt surface area in the full-scale WTP HLW melter. The operation of the ADS pump is controlled from the melter computer control system. The ADS pump works by opening the pump reservoir to the feed tank using a double-acting air cylinder and mechanical link to actuate the poppet. The reservoir is filled with slurry by gravity. After sufficient time is allowed to fill the reservoir (a few seconds), the poppet is toggled to close the reservoir to the tank and open the transfer line. After a desired delay time (dependent on the desired feed rate) the reservoir is pressurized with air to transfer the slurry (about 1.6 liter/shot) to the melter. This cycle is repeated at the rate required to provide the desired feed rate.

When necessary, a backup system is used to introduce feed into the melter with an air operated diaphragm (AOD) pump system that simulates the pulsed feeding action of an ADS pump. The recirculation loop extends to the top of the melter where feed is diverted from the recirculation loop into the melter through a Teflon-lined feed line and water-cooled feed tube. Two computer-operated pinch valves, one on the feed line and one on the recirculation loop, are activated in a timed sequence to introduce feed into the melter at the desired rate. The feed rate is regulated by adjusting the length of each pulse, the time between each pulse, and the pressure applied to the recirculation loop.

#### 1.4.2 Melter System

The DuraMelter 1200 (DM1200), which is the HLW Pilot Melter, was used for these tests. Cross-sectional diagrams of the melter illustrating the discharge chamber and electrode configuration are provided in Figures 1.3 and 1.4. The DM1200 is a Joule-heated melter with Inconel 690 electrodes and thus has an upper operating temperature of about 1200°C. The melter

shell is water-cooled and incorporates a jack-bolt thermal expansion system. The footprint of the melter is approximately 8 ft. by 6.5 ft. with a 4 ft. by 2.3 ft. air-lift discharge chamber appended to one end; the melter shell is almost 8 ft. tall. The melt surface area and the melt pool height are approximately 32 percent and 57 percent, respectively, of the corresponding values for the fullscale HLW melter. The discharge riser and trough are full-scale to verify pouring performance. Other aspects of the discharge system are also prototypical such as the chamber ventilation scheme. The glass contact refractory is Monofrax<sup>®</sup> K-3 while the plenum area walls are constructed of Monofrax<sup>®</sup> H refractory. The surface of the glass pool is 34" by 54" with a nominal glass depth of 25". The resultant melt volume is approximately 45,000 cubic inches (735 liters), which represents a glass tank capacity of more than 1.7 metric tons of glass. However, since the typical operating glass level is closer to 29 inches, the effective glass volume during testing is actually about 849 liters, giving an inventory of about 2.0 metric tons. The DuraMelter<sup>™</sup> 1200 is fitted with one pair of electrodes placed high on opposite walls of the melter as well as one bottom electrode. The side electrodes are 11" by 34" giving an electrode area for the pair of about 750 sq. in. Depending on the glass level, the plenum space extends about 33" to 36" above the melt surface resulting in a plenum volume ranging from about 43 to  $46 \text{ ft}^{3}$ .

The single-phase power supply to the melter electrodes (250 kW design power) is derived from the DuraMelter 1000 transformers by wiring them in parallel and using a single large silicon controlled rectifier. Current can be passed either from the side electrodes to the bottom electrode or between the two side electrodes only, by rearranging jumpers; only side-to-side operation was used for the present tests. Programmable process controllers are installed and can be used to control temperature or power. The melt temperature is controlled by configuring the process controller to maintain constant power and adjusting the power set-point as needed to maintain the desired operating temperature. Alarms can be set to detect out-of-range temperatures or power in the melter. Backup process controllers are installed to be used in case of failure of the main controllers. The entire system is supported by a back-up generator that is tripped on in the event of a power outage.

The DuraMelter 1200 has several other features. The lid refractory is prototypic and also includes a two-piece construction, which simulates the seam needed for the LAW lid that was planned to be fabricated in three pieces. Nozzles are provided for the off-gas film cooler, a standby off-gas port, discharge airlift, along with 11 ports available for top-entering bubblers, start-up heaters and other components as needed. In addition, a bubbler arrangement is installed in the bottom electrode with the objective of developing permanent bubblers for possible use on future melters. For the present tests the optimum bubbler configuration established during previous tests with HLW simulants [6] consisting of two double-outlet, top-entering bubblers was used, located in positions to mimic conditions in the WTP HLW melter. Figure 1.5 shows a schematic of the prototypical double-outlet bubbler design that was based on the combination of the results from these DM1200 tests [6] and room-temperature tests that were performed in a transparent fluid simulating the properties of the glass melt [47]. These bubblers have outlets 8 inches apart and were placed on the melter floor. The orientation of the bubblers in the melter, as shown in Figure 1.6, results in one of the bubbling outlets being 11.3 inches from the feed tube.

#### 1.4.3 Off-Gas System

The melter and entire off-gas treatment system are maintained under negative pressure by two Paxton external induced draft blowers. This negative pressure is necessary to direct the gases from the melter to the prototypical off-gas system. The off-gas treatment system, shown schematically in Figure 1.7, consists of a submerged bed scrubber (SBS); a wet electrostatic precipitator (WESP); a high-efficiency mist eliminator (HEME), a high-efficiency particulate air (HEPA) filter; a thermal catalytic oxidation unit (TCO); a NO<sub>x</sub> removal system (SCR); a caustic packed-bed scrubber (PBS); and a second HEME. Note that the PBS and the second HEME are not part of the WTP off-gas train, which effectively ends at the SCR. The HEME is used to limit entrained particle carryover into the balance of the VSL ventilation system. The system can be functionally divided into four subsystems:

- <u>Particulate Removal</u>: Components from the SBS to the HEPA serve to remove essentially all of the particulate from the gas stream with an estimated removal efficiency of greater than 99.9999% for particles greater than 0.3  $\mu$ m in size. In the WTP facility, this provision serves to segregate the radioactive from the non-radioactive components in the system for maintenance and handling purposes.
- <u>VOC Control/Acid Gas</u>: The TCO unit is designed to oxidize any hazardous organics that are present in the off-gas stream. This is followed by a SCR to remove NO<sub>x</sub> gases and a PBS to remove remaining acid gases.
- <u>Stack System</u>: The emergency/bypass exhaust system, which includes a second HEPA, and the primary off-gas system both feed into the building stack system for exhausting to the atmosphere.
- <u>Liquid Processing</u>: Components including the water spray lines, liquid sampling and water storage tanks, as well as the effluent evaporator, function to sample and process the system liquids for recycle or discharge.

With minor exceptions, the DM1200 off-gas system processing sequence follows the design for the full-scale WTP HLW melter system, except for cooling of the off-gas stream discharged from the SCR unit (which is present in the WTP off-gas train, but absent in the DM1200 system). Per WTP direction, the SBS unit that was used for previous DM1200 testing was modified in early 2004. Installation of the new system was completed in March 2004 and that unit was used for the present tests. The changes were implemented to reflect modifications to the WTP SBS design that have taken place since the original DM1200 unit was installed. These modifications included changes to the diffuser plate design, down-comer jacket and connection to the diffuser plate, bed diameter, bed packing materials, cooling coils, and liquid overflow level.

Initial quenching of the melter exhaust gas stream is effected by the film cooler. Immediately upstream of the film cooler is the injection point for control air, which is used to regulate melter pressure. The gas entering the balance of the off-gas system is at a temperature of about 250 to 350°C and a flow rate of about 100-250 scfm, of which about 10-80 scfm is water vapor. The off-gas is then rapidly quenched by direct liquid water contact in the SBS, which also effects removal of most of the larger particulates. The piping between the film cooler and SBS has a high superficial gas velocity to minimize particulate deposition. The gas stream leaving the SBS is at a low temperature (typically between 40-50°C). Further mist and particulate removal is effected in the WESP, HEME and HEPA. The TCO and SCR follow the particle removal components and serve to destroy organic compounds and nitrogen oxides. These two units were off-line during the present tests due to the low concentrations of these components in the exhaust stream. Finally, the PBS provides acid gas removal. Water sprays are located in the WESP, PBS, and facility HEME to wash down deposits and dissolved species into their respective collection sumps from which they can be sampled. The system components are fabricated from corrosion resistant materials, including AL6XN and 316L stainless steel, and various plastics in less demanding locations. There are extensive provisions for sampling both the gas and liquid streams throughout the system in order to collect mass balance information and removal efficiency data for each treatment stage.

The off-gas system maintains the melter plenum under slight negative pressure, typically about -5 in. W.C. The plenum pressure is controlled by means of an air injection system that introduces a controlled air flow into the off-gas jumper just after the film cooler. The air is supplied by a blower through a diverter valve. The setting of the diverter valve, and therefore the air flow rate, is controlled by a process controller that responds to the signal from a melter pressure transducer. When the plenum pressure becomes more positive, the air injection flow rate is decreased, which tends to restore the pressure to the set-point. Conversely, the flow rate is increased when the plenum pressure becomes more negative.

#### **1.5 Feed Sample Analysis**

Feed samples were taken directly from the feed recirculation line during each test. Feed samples were poured into a platinum/gold crucible that was placed into a programmed furnace for drying and fusion to form a glass. The glass produced from this fusion was ground to less than 200 mesh and sealed in 20-ml vials for subsequent analysis by x-ray fluorescence spectroscopy (XRF), or by acid digestion followed by direct current plasma - atomic emission spectroscopy (DCP-AES) on the resulting solution. The feed samples were also characterized for their rheological properties, density, pH, water content, and glass yield.

#### **1.6 Glass Product Analysis**

The glass product was discharged from the melter into either 5-gallon steel pails (DM100) or 55-gallon drums (DM1200) periodically using an air-lift system. The discharged product glass was sampled at the end of each test by removing sufficient glass from the top of the cans for compositional analysis and secondary phase determinations. In addition, the Product Consistency Test (PCT) and Toxicity Characteristic Leaching Procedure (TCLP) were performed on samples of the glass product from the DM100 melter tests. Prior to those tests, the PCT and

TCLP were also performed on the crucible melt compositions that were selected for the melter tests to ensure their compliance with the present WTP contract requirements. All of these procedures are routinely conducted at VSL and, therefore, standard operating procedures (SOPs) are in place.

Sample preparation for chemical analysis typically involves size reduction and sieving. All samples were subjected to XRF to determine the concentration of all elements except boron and lithium. A series of National Institute of Standards and Technology (NIST) reference materials were used for confirmation of the XRF data. Boron and lithium were determined by total acid dissolution of ground glass samples in HF/HNO<sub>3</sub> and subjecting the resulting solutions to DCP-AES analysis.

#### 1.6.1 Viscosity

The melt viscosity,  $\eta$ , was measured using a Brookfield viscometer. Measurements are performed in the temperature range of 950-1250°C and the data are interpolated to standard temperatures using the Vogel-Fulcher equation:  $\ln \eta = [A/(T-T_o)]+B$ , where A, B, and T<sub>o</sub> are fitting parameters. The equipment is calibrated at room temperature using standard oils of known viscosity and then checked at 950-1250°C using a NIST standard reference glass (SRM 711). Both precision and accuracy of the viscosity measurements are estimated to be within ±15 relative%.

#### **1.6.2** Electrical Conductivity

The electrical conductivity,  $\sigma$ , of each glass melt was determined by measuring the resistance of the glass melt as a function of frequency using a calibrated platinum/rhodium electrode probe attached to a Hewlett-Packard model 4194A impedance analyzer. Measurements are performed over similar temperature ranges to those employed for the melt viscosity measurements. The results are analyzed and modeled to obtain the DC electrical conductivity. The electrical conductivity data are then interpolated to standard temperatures using the Vogel-Fulcher equation:  $\ln \sigma = [A/(T-T_o)] + B$ , where A, B and T<sub>o</sub> are fitting parameters. Estimated uncertainties in the electrical conductivity measurements are  $\pm 20$  relative%.

#### **1.6.3** Product Consistency Test (PCT)

The product consistency test (PCT; ASTM C 1285) is used to evaluate the relative chemical durability of glasses by measuring the concentrations of the chemical species released from 100-200 mesh crushed glass (75-149  $\mu$ m) to the test solution (de-ionized water in this case). PCT tests on the HLW glasses are performed at 90°C, in accordance with the current WTP contract requirement. The ratio of the glass surface area to the solution volume for this test is about 2000 m<sup>-1</sup> (typically, 10 g of 100-200 mesh glass is immersed in 100 ml deionized water). All tests are conducted in triplicate, in 304L stainless steel vessels, and in parallel with a standard glass included in each test set. The internal standard is the Argonne National Laboratory – Low

Activity Waste Reference Material (ANL-LRM) reference glass [48] and/or the Defense Waste Processing Facility (DWPF)-Environmental Assessment (EA) glass, both of which have undergone round-robin testing. The leachates are sampled at predetermined times, the first of which is seven days. One milliliter of sampled leachate is mixed with 20 ml of 1M HNO<sub>3</sub> and the resulting solution is analyzed by DCP-AES; another 3 ml of sampled leachate is used for pH measurement.

#### **1.6.4** Toxicity Characteristic Leaching Procedure (TCLP)

The TCLP was performed at VSL using SW-846 Method 1311, which employs leaching of crushed glass (< 3/8") in a sodium acetate buffer solution for 18 hours at 22°C with constant end-over-end agitation. A mass of about 100 grams of glass is leached in 2 liters of TCLP extract, according to the extraction method for non-volatiles. The surface area to volume ratio for this test is about 20 m<sup>-1</sup>, which is about two orders of magnitude lower than that in the PCT. The leachates are analyzed by DCP-AES according to VSL standard operating procedures.

#### **1.6.5** Secondary Phases

Secondary phases in the glass samples were determined by optical microscopy and scanning electron microscopy coupled with energy dispersive x-ray spectroscopy (SEM-EDS). Secondary phases due to crystallization and phase separation can be identified using these methods. Quantitative determination of the amount of crystals in glass samples were made by SEM in conjunction with image analysis.

#### SECTION 2.0 WASTE SIMULANT AND BASE GLASS COMPOSITION

#### 2.1 Waste Simulants

The waste stream compositions previously provided by DOE are given in Table 2.1 on an oxide basis [5]. The work described in this report focused exclusively on the aluminum limited waste stream in response to the comparatively low glass production rates achieved with this waste stream in the earlier tests [2]. Actual Hanford HLW tank wastes are aqueous solutions with suspended solids and dissolved salts including hydroxides, nitrates, nitrites, halides, and carbonates. For the purpose of the previous [2] and present work, the concentrations of the volatile components (i.e., carbonate, nitrite, nitrate, and organic carbon) are assumed to be similar to those found for the AZ-102 HLW waste [21]. With the waste composition defined, formulation of the HLW waste simulant proceeds in a straightforward fashion. In general, oxides and hydroxides are used as the starting materials, with slurry of iron (III) hydroxide (13% by weight) as one of the major constituents. Volatile inorganic components are added as the sodium salts, whereas organic carbon is added as oxalic acid. Finally, water content was adjusted to target a glass yield of 500 g of glass per liter of feed. Two waste simulants were employed, with the only difference being the replacement of aluminum oxide by aluminum hydroxide in order to investigate the effects of variations in the form of aluminum in the Hanford HLW streams on feed properties and processing rates. The compositions of the waste simulants with aluminum oxide and aluminum hydroxide, formulated to produce 100 kg of waste oxides, are given in Tables 2.2 and 2.3, respectively.

#### 2.2 Base Glass Formulation

Several glass compositions spanning a range of waste loading were previously evaluated for suitability for the high aluminum simulated waste [2]. Based on those results, a glass with 45 wt% waste loading, HLW-E-Al-27, which employed moderate additions of alkali and alkaline earth oxides (Na<sub>2</sub>O, Li<sub>2</sub>O, and CaO), B<sub>2</sub>O<sub>3</sub>, and SiO<sub>2</sub>, was selected as providing the best combination of high waste loading and glass and melt properties [2]. Table 2.4 presents the composition of this glass and the measured properties of the crucible glass, which meet all of the processing and product quality requirements. The glass contains 23.97 wt% Al<sub>2</sub>O<sub>3</sub>, which is more than two times the WTP contract minimum for Al<sub>2</sub>O<sub>3</sub> (11 wt %). All of the measured processing parameters are within acceptable ranges. The PCT leach rates are over an order of magnitude lower than those of the DWPF-EA glass and the TCLP leachate concentrations are all below the WTP delisting limits. Sodium aluminosilicate formation (e.g., nepheline) on heat treatment (especially canister centerline cooling (CCC) heat treatment) is a known concern with high-aluminum formulations and was the waste-loading-limiting factor in the formulation of HLW-E-Al-27. The selected glass produced very little crystallization (~1.9 vol%) after CCC heat treatment and the PCT response of the heat treated glass also meets the PCT requirements by wide margins.

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Based on the results from the previous work [2], the HLW-E-Al-27 glass formulation was used as the starting point for the present work. Although the HLW-E-Al-27 glass formulation achieved high waste loadings, the results of DM100 melter tests showed that the glass production rates were lower than desired [2]. Therefore, the HLW-E-Al-27 glass formulation was used as the starting point for the glass formulation work in the present study with the objective of achieving increased glass production rates while maintaining the high waste loading. That work is described in Section 3. In addition, this glass formulation was used in DM100 melter tests to assess the effects of variations in the form of aluminum in the waste simulant by replacing aluminum oxide, which was used in the previous work [2], by aluminum hydroxide. The melter feed formulation for that case is presented in Table 2.5. The results of that and other DM100 melter tests are presented in Section 4.

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#### SECTION 3.0 GLASS FORMULATION

#### 3.1 Introduction

The principal objective of the glass formulation and small scale melt rate testing work was identifying glass compositions that maximize processing rates for the Al-limited HLW waste stream specified by ORP. The previous tests [2] with the Al-limited waste steam demonstrated substantial increases in waste loading; however, the DM100 production rates were significantly lower than the rates obtained for typical iron limited wastes and, at the nominal bubbling rate and temperature, were also below the WTP target rate of 800 kg/m<sup>2</sup>/d (corresponding to 3 MT of HLW glass per day per HLW melter). The goal of the current work was therefore to retain high waste loadings and acceptable glass properties for the aluminum limited waste while increasing the glass production rate through the manipulation of the glass formulations and glass forming additives.

In view of the above objectives, melt rate determination was a critical parameter in the glass formulation effort. Melt rates are affected by many factors and are the net result of a combination of many complex processes that occur in the feed-to-glass conversion process. The cold cap that forms when liquid slurry feed is introduced onto the melt surface is highly stratified in both temperature (from  $\sim 100^{\circ}$ C on the top to  $\sim 1150^{\circ}$ C at the bottom) and composition. In this region, water and other volatiles are evaporated, salts are decomposed and melted, various transient phases are formed and consumed, and finally, new glass is formed. In addition, these reactions are governed by the substantial flows of heat and mass through this region. In view of this complexity, continuous melter tests provide the most reliable means of capturing these phenomena and, therefore, for reliably determining melt rates; furthermore, such tests become increasingly more reliable as the melter scale and test duration are increased. However, because of the cost and schedule implications, such tests are not well suited for integration into a glass formulation development program in which a large number of formulation variations need to be assessed. For this reason, in the present work we have employed two types of melt rate screening tests that permitted rapid evaluation of relative melt rates. The results from these tests were used to refine the glass formulations and down-select a subset of glasses for DM100 melter testing. The results of the DM100 tests were then used to select one formulation for pilot scale testing on the DM1200 melter system.

The two melt rate screening tests used in this work were designed to capture many of the essential features of the feed-to-glass conversion process, albeit in a necessarily simplified fashion. In the Vertical Gradient Furnace (VGF) tests, dried feed is placed on top of pre-melted glass in a crucible and subjected to a vertical temperature gradient for predetermine times (see Section 3.2.2). The crucible is then quenched, sectioned, and examined to determine the nature and extent of the feed conversion process. In the DM10 Feed Consumption (DFC) tests, an aliquot (~1 kg) of slurry feed is introduced onto the surface of the DM10 melter and the time to consume the feed is determined by a combination of visual observation and the change in

plenum temperature (see Section 3.4). Both tests are well suited to quickly and efficiently screening relatively large numbers of composition variations. The tests successfully identified formulations with significantly improved melting rates and the results were subsequently confirmed in continuous melter tests at the DM100 and DM1200 pilot melter scale.

This section presents descriptions of the glass formulation approach, the test methods employed, and the results and conclusions from the melt rate screening tests and glass formulation development work.

#### **3.2** Experimental Methods

The results from the earlier work on the aluminum limited waste [2], and specifically the glass formulation selected in that work (HLW-E-Al-27), provided the starting point for the present glass formulation effort. An iterative approach was employed in which composition modifications were designed that were intended to improve melt rates, crucible melts of those formulations were prepared, and characterization data were collected. The results were then analyzed and used to design additional formulations for testing. To improve efficiency, glass characterization was conducted in stages such that glasses that failed any processability or product quality requirement were not subjected to further testing. All glasses were tested for phase behavior, both as-melted and after heat treatment, since that was expected to be one of the most limiting constraints. Acceptable glasses were then subjected to testing with respect to PCT, melt viscosity, melt electrical conductivity, and TCLP. Glasses that met these requirements were then subjected to melt rate screening tests. In parallel, a series of DFC melt rate screening tests were performed on base glass formulations in which simple one or two component variations were made in order to collect information on component effects on melt rate (see Section 3.4). These results were also factored into the glass formulation design effort. The experimental methods that were employed in this program are described in the following sections.

#### 3.2.1 Crucible Melts

Crucible melts of each glass were fabricated at VSL using reagent grade chemicals, mostly oxides and carbonates. Glass preparation began with a batching sheet that provided information on the required starting materials. The information included the chemicals needed, identification of the chemicals according to the vendors and catalog numbers, the associated purity, together with the amount required to produce a given amount of glass. Chemicals were weighed and batched according to the batching sheets. After the starting materials were weighed and batched, a blender was used to mix and homogenize the starting materials before they were loaded into platinum/gold crucibles that were engraved with individual identification numbers. The loaded platinum/gold crucible was placed inside a Deltech DT-28 (or DT-29) furnace, the heating of which was controlled by a Eurotherm 2404 temperature controller. The melting temperature was 1200°C, at which the melt was kept for 2 hours. Mixing of the melt was accomplished mechanically using a platinum stirrer, beginning 20 minutes after the furnace

temperature reached 1200°C and continuing for the next 90 minutes. The molten glass was poured at the end of 120 minutes onto a graphite plate to cool before recovery.

All of the as-melted glasses were inspected for signs of phase separation and completeness of melting; secondary phases were analyzed by SEM-EDS and optical microscopy, as described in Section 1.6.5. The composition was checked XRF and DCP analysis, as described in Section 1.6.

Selected glasses were subjected to heat treatment for 70 hours at 950°C and below (900, 850 and 800°C). Glass samples (about 5 grams each) were heat-treated in platinum, platinumgold, or platinum-rhodium crucibles (5 ml) at a pre-melt temperature of 1200°C for 1 hour, followed by heat treatment at the prescribed temperatures. At the end of the heat-treatment period, the glass samples were quenched by contacting the crucible with cold water. This quenching freezes in the phase assemblage in equilibrium with the melt at the heat-treatment temperature. The types and amounts (vol%) of crystalline phases were determined by SEM-EDS.

Selected glasses were subjected to canister centerline cooling (CCC) heat treatment according to the WTP HLW CCC temperature profile [13]. As in the case of isothermal heat-treatment, the glass samples in platinum crucibles were maintained at a pre-melt temperature of 1200°C for 1 hour before initiation of the CCC treatment. The samples recovered after CCC heat treatment were subjected to SEM-EDS examination.

Selected glasses were also characterized with respect to their melt viscosity and electrical conductivity, and PCT and TCLP leach testing, as described in Sections 1.6.1 - 1.6.4, respectively. The PCT was performed on both quenched samples and glass samples that had been subjected to CCC heat treatment.

Tables 3.1 - 3.3 provide the identifications, compositions and properties of the glass formulations and crucible melts that were tested in this work.

#### **3.2.2** Vertical Gradient Furnace

As described above, the cold cap in a continuously fed melter is subject to a large temperature gradient in the vertical direction. This gradient can drive heat and mass flows and leads to variations of reaction rates at different heights in the cold cap; the gradient is therefore a potentially significant factor in determining the melt rate. The design of the Vertical Gradient Furnace (VGF) melt rate screening test emphasizes the large temperature gradient in the vertical direction across the cold cap.

Figure 3.1 is a schematic diagram the VGF setup. The temperature gradient inside the VGF is maintained by two separate sets of heating elements, both of which are arranged in cylindrical form and aligned along their axis. The inner heater is set at 1150°C, which is the nominal temperature of the glass pool, and the ambient heater is set at 600°C, which is similar to the melter plenum temperature. A ceramic crucible (4 inches tall) is used to contain the reacting melter feed. The temperature gradient in the loaded furnace is shown in Figure 3.2. For a typical

feed conversion test, a sample of dried melter feed equivalent to 20 grams of glass is introduced into the ceramic crucible, which already contained about 10 grams of pre-melted glass of the same composition that had been preheated in the inner heater. Feed reactions under the controlled temperature gradient are allowed to continue for the designated test duration and then stopped by rapid cooling in room temperature air. The top surface, and the cross section (by sectioning the crucible) of the reacted feed are then inspected and photographed. Samples of the partially reacted feed are taken for further characterization by SEM-EDS, X-ray diffraction (XRD), and XRF. The composition of the feed was confirmed by XRF analysis of samples that were fused at 1150°C.

All of the melter feed samples were prepared using the same HLW simulant and preparation methods that were used for the melter tests, as described in Section 2. The only difference was that samples used for VGF tests were dried, crushed, and sieved, before use.

The VGF results were used to evaluate the melt rate on a relative scale using the degree of the melting that had occurred, the structure of the feed materials that were undergoing reaction and transformation, and the conversion progression with time. A numerical ranking of relative melt rate was assigned based on calibration tests using feeds whose melt rates had been determined previously in DM100 melter tests.

#### **3.3** Results and Discussion

The identifications, compositions and properties of the glasses designed and tested in the present work are listed in Tables 3.1 - 3.3. The feeds that were subjected to VGF testing are listed in Tables 3.1 and 3.4. These include the feeds corresponding to new glass formulations, several feeds from previous work that were used for calibration, and several feeds that are based on the earlier HLW-E-Al-27 formulation [2]. The results of the glass formulation, characterization, and melt rate tests are presented together below in groups according to the changes made in key glass formers with respect to the earlier HLW-E-Al-27 glass [2].

#### 3.3.1 VGF Calibration Tests

As part of the system calibration and method development work, six melter feeds of known glass production rate (as determined in DM100 tests at 1150°C with nominal bubbling) were tested in the VGF. The DM100 glass production rates, glass compositions, and the feed recipes were reported previously [2, 24, and 25]. The same feed samples as those tested in the DM100 melter were used for the VGF tests. Two of the feeds were based on iron-limited glasses [24, 25] and the remaining four were based on Cr-, Bi-, Al-, or Al+Na-limited HLW streams [2].

Figure 3.3 shows the top views and the cross sections of the six feed samples after VGF tests. It is apparent that the six melter feeds exhibit a wide range of feed conversion progress, from virtually completely melted (HLW-E-CrM) to development of a complete cold cap "dome" that has lost contact with the underlying glass and shows little sign of reaction on the surface

(HLW-E-Al-27). Feed of HLW-E-Bi-6 shows greater melting progress than HLW-E\_Al-27, but exhibits a partial dome structure. Similar to HLW-E-CrM, the two iron-limited feeds (Matrix 1-B1 and Matrix 2-9) and the Al/Na-limited feed (HLW-E-ANa-22) display only traces of the once-present dome, with remnants of rims folded on the side wall and undergoing melting. On a relative scale, using the slowest (HLW-E-Al-27) at one end and the fastest (HLW-E-CrM) at the other end, the degree of conversion of the dried melter feed materials after 30 minute VGF experiments can be divided into several intermediate stages based on the top views and cross sections. Table 3.5 lists the key observations for each stage and their assigned ranking on a relative scale. It should be emphasized that these assignments are simply rankings; a melt rate ranking of 2 should not be interpreted to mean two times faster than a melt rate ranking of 4.

The results for the six feed samples in terms of their ranking values are plotted against the glass production rates of the same melter feeds as by determined in DM100 tests in Figure 3.4. A more or less linear trend is defined by the two extreme feed samples and three intermediate samples. It is not clear why the result for HLW-E-ANa-22 feed falls off this trend. However, the DM100 rate of 400 kg/m<sup>2</sup>/d is unusually low and, in fact, is the lowest rate ever measured under these test conditions. Furthermore, the increase in the DM100 rate result when the temperature was increased from 1150°C to 1175°C was unusually large (125% vs. the more typical ~25%) [2]; based on the measured rate at 1175°C and the more typical change with temperature, a rate of about 720 kg/m<sup>2</sup>/d at 1150°C would have been expected, which would be much more consistent with the VGF trend. It is possible that other atypical effects were involved in the unusually low rates in the 1150°C DM100 test. Further tests would be useful to clarify this issue. Nevertheless, the general trend proved to be adequate to render the VGF test a useful screening tool for the development of glass and feed formulations with improved melt rates. This is particularly so when the VGF is used in combination with the DFC screening test results, as was the case in the present work.

#### 3.3.2 Component Effects on the HLW-E-Al-27 Formulation

The high waste loading glass (HLW-E-Al-27) developed for the aluminum limited waste in the earlier work [2] was used as the starting point for the melt rate improvement work. During DM100 tests [2] the phenomenon of "shelving" or "bridging" in the cold cap region was observed. The slurry feed had a tendency to develop a shelf-like structure of hardened feed materials that could attach to the side walls of the DM100 melter. Without intervention, the shelf could extend and form a bridge above the glass line. The separation between the melter feed and the hot glass melt can cause the melt rate to decrease. It is interesting to note that a similar phenomenon was also observed in the partially reacted melter feed of HLW-E-Al-27 in VGF tests. As shown in Figure 3.5, a closed dome-like structure developed after 30 minutes in the VGF. The partially molten inner lining of the dome apparently provides a sufficient seal to allow the gas from the feed materials closer to hot glass surface to cause the cold cap to balloon upwards and loose contact with the underlying glass melt. As the dome expands, the temperature drops because of the vertical temperature gradient and the dome hardens. Feed materials on the top of the dome are relatively insulated from the hot glass surface by the dome and intervening gas. As shown in the top view, the granular feed materials do not exhibit much sign of reaction/transformation after 30 minutes in the VGF. Eventually, sufficient heat is delivered to melt away part of the dome and a vent emerged after 45 minutes in the VGF. As the reaction/melting continues, the dome structure softens and collapses to the side wall of the ceramic crucible. The similarity of the evolution of the dome-like structure observed in a series of VGF experiments and the cold cap behavior in the DM100 tests suggests that the VGF tests indeed capture some of the key features of the melt rate determining processes.

The chemicals used in Al-limited HLW simulant in the previous melter test are listed in Table 3.6. The typical glass former additives are lithium and sodium carbonates, borax or boric acid, and simple oxides such SiO<sub>2</sub> and ZnO. The development of a dome-like structure in VGF tests suggests a need for more lower-melting or more reactive components in the HLW-E-Al-27 glass/feed formulation. The original feed formulation used aluminum oxide and borax [2]. The selection of aluminum oxide was based a conservative (in terms of melt rate) interpretation of the form of aluminum in the waste; more likely forms are the hydroxide (gibbsite), oxy-hydroxide (boehmite), and various silicates. Therefore tests were performed with the feed of HLW-E-Al-27 reformulated using aluminum hydroxide and/or boric acid (sodium carbonate was used to compensate the sodium difference). Three slurry feeds of the same target glass composition HLW-E-Al-27 were prepared with different combinations of Al(OH)<sub>3</sub> and borax, Al<sub>2</sub>O<sub>3</sub> and boric acid, and Al(OH)<sub>3</sub> and boric acid (Table 3.1). The feed materials were tested in the VGF under the same experimental conditions for 45 minutes (Figure 3.6). The results suggest that both substitution (aluminum hydroxide for aluminum oxide or boric acid for borax) improve the melt rate. The best result was obtained for the feed that includes both aluminum hydroxide and boric acid. The VGF feed conversion results strongly suggest that boric acid should be used instead of borax as the primary boron source.

In summary, boric acid appears to be preferred as the primary boron source. Also, in view of the variability and uncertainty with respect to the form of aluminum in the waste, it is reasonable to include both aluminum oxide and aluminum hydroxide versions in the melt rate evaluation to provide reasonable bounds. In the subsequent VGF tests, all of the melter feeds for the new glass formulations were prepared using boric acid and  $Al_2O_3$ . The two formulations that were later selected for DM100 melter tests were also tested using the corresponding  $Al(OH)_3$  versions.

#### **3.3.3 Group 1 Glass and Feed Formulations**

The glass formulation and feed test results are presented in groups according to changes made in key glass formers with respect to the baseline glass HLW-E-Al-27.

After reviewing the chemistry of HLW-E-Al-27, 5 wt% CaO added as a glass former was suspected as a likely a contributor to the overall slow melt rate observed in the DM100 melter tests. The calcium addition was found to be effective in suppressing spinel formation, which allowed higher waste loadings to be achieved. Therefore, many of the new glass formulations in this work investigated the phase behavior of high-Al glasses at different levels of CaO (zero is not possible because the Al-limited waste contains small amounts of CaO) and their melt rate as

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determined by VGF. In the first variation, HWI-Al-1 has the same 45 wt% waste loading as the baseline glass but with the CaO additive replaced by  $B_2O_3$  and  $Li_2O$ ; this glass contained minor amounts of crystalline phase (spinel) in the quenched glass (Tables 3.2 and 3.3). More spinel crystallized after heat treatment at temperatures from 950 to 800°C (Table 3.3). Although the substitution of  $B_2O_3$  and  $Li_2O$  for CaO increased crystallization near the glass melting temperature, it improved the melt rate, as shown in Figure 3.7 and Table 3.3.

At a reduced waste loading of 40 wt%, glass HWI-Al-9 was homogeneous as-melted and developed little crystallization after heat treatment at temperatures from 950 to 800°C and under CCC conditions (Tables 3.2 and 3.3). In this case, the reduced waste loading coupled with higher sodium contributed to the suppression of spinel crystallization. The results from VGF testing indicate similar melt rates for feeds of HWI-Al-9 and HWI-Al-1. HWI-Al-10 was formulated in a manner similar to HWI-Al-9 but with minor adjustments in B<sub>2</sub>O<sub>3</sub> and Li<sub>2</sub>O concentrations. Since very similar heat treatment result were obtained (Tables 3.2 and 3.3), no further tests were performed on this glass.

The results for the three glasses investigated in this group highlight one of the challenges involved in modification of the baseline glass formulation. Although the melt rate for HWI-A1-9 by VGF is considerably faster than for the baseline feed, the reduction in waste loading needed to suppress crystallization is undesirable. In the four groups of formulations presented below, other glass forming additives are investigated for their roles in achieving the overall goal of improving the melt rate while suppressing crystallization of spinels, without reducing the waste loading.

#### 3.3.4 Group 2 Glass and Feed Formulations

Two glasses, HWI-Al-2 and HWI-Al-3, were formulated with an additional one percent of  $P_2O_5$  and at similar levels of  $Na_2O$  and  $B_2O_3$  to HWI-Al-9. Both glasses showed signs of phase separation in the quenched samples (Tables 3.2 and 3.3) and therefore no further testing was performed.

#### **3.3.5** Group 3 Glass and Feed Formulations

In this group, CaO as an additive in HLW-E-Al-27 was partially or completely replaced by MgO to produce glasses HWI-Al-5 and HWI-Al-7, respectively (Tables 3.2 and 3.3). Spinel crystallization near the glass melting temperature that was observed in many low CaO glasses in this work was not present in either of the MgO-added glasses (Table 3.3). However, more crystalline phases developed after heat treatment at temperatures from 950 to 800°C and under CCC conditions (Table 3.3). Similar to the baseline feed, the melt rates of the both formulations in VGF tests were rather slow (Figure 3.8 and Table 3.3). Although a well developed dome structure was not present, the MgO-added feeds appeared to soften but were resistive to further melting. In view of the increased crystallization and limited indication of improved melting, substitution of MgO for CaO was not further pursued.

#### 3.3.6 Group 4 Glass and Feed Formulations

The effect of  $K_2O$  was investigated in this group. One of the motivations for introducing  $K_2O$  into the glass formulations is to suppress the formation of iron-chromium spinel near the glass melting temperature, which was employed in previous work with high-iron HLW [1]. In addition, the results from DFC melt rate screening tests (Section 3.4) suggested some improvement in melt rates with additions of  $K_2O$ .

Three glasses, HWI-Al-4, HWI-Al-6 and HWI-Al-8, were formulated with different levels of  $K_2O$  (4, 6, and 8 wt% respectively; Table 3.2) largely in place of CaO. However, all three glasses showed clear signs of secondary phases in the quenched samples (Table 3.3). Additional Li<sub>2</sub>O and B<sub>2</sub>O<sub>3</sub> in HWI-Al-20 failed to suppress the formation of secondary phases, although the VGF melt rate was significantly improved compared to the baseline formulation (Tables 3.2 and Table 3.3; Figure 3.9).

At a similar level of CaO as in the baseline glass but with additional  $K_2O$  in place of Na<sub>2</sub>O (HWI-Al-13, HWI-Al-14) or B<sub>2</sub>O<sub>3</sub> (HWI-Al-15), crystallization of spinel in the quenched glasses was reduced considerably for HWI-Al-13 and HWI-Al-14 and little changed for HWI-Al-15 (Tables 3.2 and 3.3). HWI-Al-13 displayed increased crystallization of spinel and apatite after heat treatment at temperatures from 950 to 800°C (Table 3.3). Reducing the waste loading in HWI-Al-17 (to 43.6 wt%) did not significantly reduce spinel crystallization (Tables 3.2 and 3.3). VGF tests on HWI-Al-13 and HWI-Al-17 showed only minor improvement from the baseline feed and presence of extensive foam layer (Figure 3.9).

In summary, addition of  $K_2O$  was not effective in suppressing spinel crystallization in formulations without CaO as an additive and did not improve melt rates at higher CaO concentrations.

#### **3.3.7** Group 5 Glass and Feed Formulations

From the results of the above tests, it is apparent that a certain level of CaO as glass forming additive is needed in order to suppress spinel crystallization near the glass melting temperature if high waste loadings are to be maintained. Five glasses in Group 5 were formulated to investigate the effect of increased boron content in glasses at two levels of CaO. The beneficial effects of higher boron content on melt rate is suggested by both VGF and DFC (see Section 3.4) melt rate screening test results.

Two glasses, HWI-Al-11 and HWI-Al-12, were formulated with increased  $B_2O_3$  at moderate amounts of CaO (2 and 3 wt%, respectively; Table 3.2). Phase separation was observed in the quenched glass samples (Table 3.3) and therefore no further characterization was performed.

Two glasses, HWI-Al-16 and HWI-Al-19, were formulated at similar CaO concentrations as in the baseline HLW-E-Al-27 glass (Table 3.2). HWI-Al-16 results from a renormalized

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mixture of the baseline glass and 3 wt% B<sub>2</sub>O<sub>3</sub> at a reduced waste loading of 43.6 wt%. HWI-Al-19 results from replacement of 3.5 wt% SiO<sub>2</sub> and 0.5 wt% CaO by 4 wt% B<sub>2</sub>O<sub>3</sub> at the same waste loading as the baseline glass (45 wt%; Tables 3.2 and 3.3). Both HWI-Al-16 and HWI-Al-19 showed little evidence of phase separation in the quenched samples. The glass samples after heat treatment at temperatures of 950 to 800°C and under CCC conditions showed limited crystallization, mostly of spinel with minor amounts of apatite (Table 3.3). The VGF melt rates were considerably faster than most of the formulations investigated in this work (Table 3.3, Figure 3.10). HWI-Al-18 was formulated with an additional 3 wt% B<sub>2</sub>O<sub>3</sub> based on HWI-Al-18 displayed similar levels of crystallization and melt rate to those of HWI-Al-16 and HWI-Al-19 (Figure 3.10, Table 3.3).

The combination of higher boron while retaining modest additions of calcium was successful in improving the VGF melting rate while controlling spinel crystallization near the glass melting temperature. All three feeds formulated using Al<sub>2</sub>O<sub>3</sub> as the aluminum source underwent transition/melting considerably faster than the baseline feed. These melt rate improvements were also supported by the results from DFC tests (Section 3.4). Since the HWI-Al-16 and HWI-Al-19 achieve only slightly lower and the same waste loadings as the baseline formulation, respectively, these were selected in consultation with ORP for DM100 melter testing. Characterization results for glasses HWI-Al-16 and HWI-Al-19 are summarized in Table 3.3. TCLP leachate concentrations are listed in Table 3.7.

#### 3.3.8 Effect of Aluminum Source on HWI-Al-16 and HWI-Al-19

As discussed earlier, the choice of Al<sub>2</sub>O<sub>3</sub> as the source of aluminum for the HLW-E-Al-27 melter tests [2] represents a conservative scenario with respect to feed processing rate. The actual forms of aluminum in the waste are likely to be much less refractory. Therefore the feeds of target glasses HWI-Al-16 and HWI-Al-19 were reformulated using Al(OH)<sub>3</sub> and subjected to VGF tests. The top and cross section views of the partially reacted feed samples after 30 minute VGF tests are shown in Figure 3.11 in comparison to their Al<sub>2</sub>O<sub>3</sub> counterparts. Overall, the results indicate that the Al(OH)<sub>3</sub> feeds tend to react and convert somewhat faster in VGF tests. The VGF rankings of the melt rates are listed in Table 3.8. Feed samples after 30 minute VGF tests were examined by SEM-EDS and XRD. Figure 3.12 shows SEM images of the crystalline phases observed on the surface of the partially reacted feed samples. Table 3.9 lists the major phases identified from analysis of the crystalline phases by SEM and XRD.

#### **3.3.9** Confirmation of VGF Melt Rate Predictions

Based on the glass formulation work and melt rate screening using the VGF and DFC tests DM100 tests were performed on HWI-Al-16 (Al<sub>2</sub>O<sub>3</sub> and Al(OH)<sub>3</sub>) and HWI-Al-19 (Al(OH)<sub>3</sub>); in addition the HLW-E-Al-27 formulation was tested with Al(OH)<sub>3</sub> to further assess the effect of aluminum source. The tests are discussed in Section 4. The results from those DM100 tests are used here to evaluate the relative melt rate predictions made based on the VGF

test results. The four additional data points were added to the original calibration plot shown in Figure 3.4 to produce Figure 3.13. As is evident from the plot, the new data are consistent with the expectations based on the original trend line, with the same single outlier. The results suggest that the VGF test provides a valuable screening tool in glass and feed formulation, which should improve as more data become available.

### **3.4** DM10 Feed Consumption (DFC) Melt Rate Tests

In view of the complexity of the feed-to-glass conversion process, a combination of small-scale tests was used to screen feed and glass composition variations with respect to projected melt rates in order to down-select the preferred compositions for subsequent melter testing. In addition to the VGF tests described above, a second test utilizes the DM10 melter to determine the relative rate at which feed is consumed into the glass melt. This test is referred to as the DM10 Feed Consumption (DFC) test. This procedure permits the evaluation of many feed compositions and additive blends in a relatively short amount of time. Data from these tests was used both to provide guidance and confirmation of the glass formulation process.

### 3.4.1 DM10 Melter

The DM10 unit is a ceramic refractory-lined melter fitted with two Inconel 690 plate electrodes that are used for joule-heating of the glass pool and a bubbler for stirring the melt. The glass product is removed from the melter by means of an air-lift discharge system. The DM10 unit has a melt surface area of 0.021 m<sup>2</sup> and a glass inventory of about 8 kg. In these tests, feed is introduced rapidly as a single charge using a peristaltic pump. For operational simplicity, the DM10 is equipped with a dry off-gas treatment system involving gas filtration operation only. Exhaust gases leave the melter plenum through a film cooler device that minimizes the formation of solid deposits. The film-cooler input air has constant flow rate and its temperature is thermostatically controlled. Consequently, under steady-state operating conditions, the exhaust gases passing through the transition line (between the melter and the first filtration device) can be sampled at constant temperature and air flow rate. The geometry of the transition line conforms to the requirements of the 40-CFR-60 air sampling techniques. Immediately downstream of the transition line are coarse particulate filters followed by conventional pre-filters and HEPA filters. The temperature of the filters is kept sufficiently high to prevent moisture condensation. An induced draft fan completes the system.

### **3.4.2** Methodology for DFC Melt Rate Tests

The DM10 is rapidly charged with a standardized amount of feed while maintaining standard operating conditions in the melter. The mass of feed used in these tests was 1 kg. Once introduction of the feed charge is complete, bubbling is increased from near zero to 1.7 lpm. This rate corresponds to the 9 lpm used for standardized melt rate testing on the DM100 (see Section 4). Visual observations of the cold cap and monitored plenum temperatures are used as indicators

of the rate of feed consumption. An example of changes in plenum temperature over a series of DFC tests is provided in Figure 3.14. The abrupt drops in plenum temperature correspond to the additions of feed and the time required for the system to re-equilibrate is an indication of time required to consume each feed charge. The plenum temperature measurements are analyzed and compared to visual observations of the cold cap to generate a melt rate index. The melt rate index reported here is the time in minutes needed to fully consume 1 kg of feed in the DM10 melter at a bubbling rate of 1.7 lpm and a nominal melt pool temperature of 1150°C. The time needed to fully consume the feed is determined through observation of the cold-cap and analysis of the changes in the plenum temperature. Results from new feed formulations are compared to results obtained from feed samples with known DM100 processing rates to provide a projected melt rate. The relationship between production rates obtained from DM100 tests and melt rate index from this method is illustrated in Figure 3.15. As is evident from the figure, the technique is especially effective at distinguishing melt rate differences at the lower melter rates that are most important in the present work.

# 3.4.3 Feed Compositions Evaluated in DFC Tests

Six different base feed compositions were evaluated using the DFC method:

- HLW-E-Al-27 glass composition; Al<sub>2</sub>O<sub>3</sub> as the aluminum source (see Table 2.2); 45 wt% oxide waste loading.
- HLW-E-Al-27 glass composition; Al(OH)<sub>3</sub> as the aluminum source (see Table 2.3); 45 wt% oxide waste loading.
- HLW-E-Al-27 glass composition; Al<sub>2</sub>O<sub>3</sub> as the aluminum source (see Table 2.2); boric acid + soda ash replacing borax; 45 wt% oxide waste loading.
- HLW-E-Al-27 glass composition; Al<sub>2</sub>O<sub>3</sub> as the aluminum source (see Table 2.2); boric acid + K<sub>2</sub>CO<sub>3</sub> replacing borax; 45 wt% oxide waste loading.
- HWI-Al-9 glass composition; Al<sub>2</sub>O<sub>3</sub> as the aluminum source (see Table 2.2); 40 wt% oxide waste loading.
- HWI-Al-9KSM glass composition; Al<sub>2</sub>O<sub>3</sub> as the aluminum source (see Table 2.2); 45 wt% oxide waste loading.

Glass compositions for each of these base feeds are provided in Table 3.10. These base feeds were tested as 1 kg batches for comparison to previously tested feeds and to each other. They were also processed at reduced masses with a variety of additives to bring the mass to 1 kg in order to evaluate the effect of these additives on melt rates. Additives evaluated on the base feeds were:

- Nitric acid and sugar
- Boric acid
- Borax
- Sodium carbonate
- Potassium carbonate

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- Calcium carbonate
- Lithium carbonate
- Silica

The combinations of base feeds and additives evaluated with the resulting glass compositions are provided in Table 3.11. It is important to note that, except for nitric acid and sugar, all of the additives result in slightly reduced waste loadings. The intent of these tests was to investigate which additives have the potential to improve melt rates and should therefore be given preference as additives in improved feed formulations. It is also worth noting that the glass compositions given in Tables 3.10 and 3.11 represent the feed compositions, not the glass pool composition. Therefore the DFC tests primarily evaluate reactions taking place in the cold cap and at the melt interface rather than effects extending throughout the melt pool.

#### 3.4.4 Results from DFC Tests

The rate of feed consumption for the six base feeds is shown in Figure 3.16. Variations based on the original HLW-E-Al-27 feed formulation showed melt rate index variations from about 60 to 100; the base feed with both forms of aluminum gave indices of about 80. Increasing the boron concentration in HWI-Al-9 and HWI-Al-9KSM reduced the melt rate index below the values for the original HLW-E-Al-27 composition, whereas simply changing the boron source increased it. The reason for the longer melt time when only the boron source was changed is unclear, particularly since HWI-Al-9 and HWI-Al-9KSM, which showed much faster melt rates, also used boric acid as the boron source. Also, previous DM100 tests evaluating the effect of boron (and sodium) source on production showed no effect on production rate [24]. However, these test results clearly indicate that feed formulations with higher boron concentrations have the potential for higher melt rates.

The effect of adding a variety of additives to four of the base feeds is illustrated in Figures 3.17 - 3.20. In these tests, 50 g of each additive was combined with 950 g of the base feed to generate 1 kg of feed that was subsequently introduced into the melter. The results indicate that some of additives have the potential to reduce melting times, some show ambiguous results, and others appear to have no effect or increase melting times. The addition of nitric acid and sugar, silica, and calcium carbonate appear to have little or no benefit with respect to reducing melt times, particularly considering the decrease in waste loading when adding silica or calcium carbonate. Lithium and sodium carbonate dramatically increase melt times in some tests, have little effect in some, and significantly decrease melt time in others. This ambiguity may be attributable to foaming on the melt pool surface generated by carbonates, which complicated cold cap observations and isolates the plenum from the hot glass pool, which slows the temperature rise. Previous tests varying elemental concentrations with a single feed composition have shown that increasing concentrations of alkali metals increase production rates [25] supporting the notion that another mechanism other than the alkali increase is responsible for the slow melt times. Potassium carbonate in all four compositions and with both forms of boron for most compositions resulted in significant decreases in melt times; hence its selection for evaluation in the glass formulation work (Section 3.3.6). Both these observation are in keeping with previous tests that show production rate increases with increasing concentrations of alkali metals and boron [25].

Additional tests were conducted with three of the base formulations to determine if progressively increasing the concentration of borax, boric acid, and potassium carbonate provides continued melt rate improvements. The results of these tests are illustrated in Figures 3.21 - 3.23. The addition of more boron in the form of borax or boric acid either maintains the initial reduction in feed consumption time from the nominal feed composition or continues to reduce the time with increasing proportions of potassium carbonate in the feed. The decreased effectiveness of potassium carbonate with increasing feed concentration may be due to foaming from the carbonate obscuring the results of the tests. However, potassium additions were not found to be effective in the VGF melt rate screening tests (Section 3.3.6).

The DFC tests results suggest that melt rate improvements are possible for high aluminum waste by appropriate feed and glass formulation changes. Based on the DFC test results, boric acid, borax, and potassium carbonate were identified as additives that have the greatest potential for increasing HLW feed processing rates whereas silica, calcium carbonate, and nitric acid plus sugar either have little effect or decrease melt rate.

#### **3.5** Summary and Formulations for Melter Testing

An integrated approach has been developed to identify feed and glass formulations with improved melt rates. The approach employs two melt rate screening tests that have been developed (VGF and DFC) and confirmatory testing in continuous melters. The VGF and DFC tests methods are sufficiently rapid and cost effective to provide for effective integration into the overall glass formulation development, testing, and optimization program.

The test results identified several additives with the potential for improving melt rates but other factors, such as increased crystallization and reduced waste loadings, also need to be considered. The HWI-Al-16 and HWI-Al-19 formulations were identified as the preferred candidates for confirmatory DM100 melter testing. Both of these formulations emphasize increased boron concentrations to improve melt rates and compensating changes to maintain other glass properties in acceptable ranges. Of these two formulations, HWI-Al-19 has the additional advantage of higher waste loading; it achieves the same 45 wt% waste loading as the baseline glass formulation but is expected to exhibit a substantial improvement in glass production rate over the baseline formulation.

The composition and properties of the HWI-Al-16 and HWI-Al-19 formulations are listed in Tables 3.12 and 3.13. The melter feed compositions for the HWI-Al-16 formulation with  $Al(OH)_3$  or  $Al_2O_3$  as the aluminum source are shown in Tables 3.14 and 3.15, respectively. The melter feed composition for the HWI-Al-19 formulation with  $Al(OH)_3$  as the aluminum source is shown in Table 3.16.

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### SECTION 4.0 DM100 MELTER OPERATIONS

Melter tests were conducted on the DM100-BL between 2/4/08 and 6/27/08. These tests produced almost two metric tons of glass from over 3600 kg of feed. Prior to each change of feed composition, the glass inventory was reduced from about 170 kg to about 100 kg in order to decrease the feeding time required to change over the composition of the melt pool. The series of eight nominally 50-hour tests were divided as follows:

- HLW-E-Al-27 with aluminum hydroxide 490 kg of glass produced
  - o 1200°C glass temperature, 9 lpm bubbling, 500 g glass per liter feed.
  - o 1150°C glass temperature, 9 lpm bubbling, 500 g glass per liter feed.
- HWI-Al-16 with aluminum hydroxide 506 kg of glass produced
  - o 1200°C glass temperature, 9 lpm bubbling, 500 g glass per liter feed.
  - o 1150°C glass temperature, 9 lpm bubbling, 500 g glass per liter feed.
- HLW-Al-16 with aluminum oxide 468 kg of glass produced
  - o 1200°C glass temperature, 9 lpm bubbling, 500 g glass per liter feed.
  - o 1150°C glass temperature, 9 lpm bubbling, 500 g glass per liter feed.
- HLW-Al-19 with aluminum hydroxide 484 kg of glass produced
  - o 1200°C glass temperature, 9 lpm bubbling, 500 g glass per liter feed.
  - o 1150°C glass temperature, 9 lpm bubbling, 500 g glass per liter feed.

Summaries of the tests are provided in Table 4.1. Attempts were made to replicate the melter configuration and operating conditions used for previous tests with HLW simulants [2, 14-17, 23-25, 44]. These conditions include a near-complete cold cap, which is between 80-95% melt surface coverage for the DM100 since a 100% cold cap tends to lead to "bridging" in smaller melters. The bubbling rate was fixed at 9 lpm and the feed rate was adjusted to maintain a complete cold cap. This use of bubbling is in contrast to some previous tests where the production rate was fixed between 1000 and 1050 kg/m<sup>2</sup>/day and the bubbling rate was adjusted to maintain the complete cold cap [14-17, 44]. The approach used in the present tests, in which the bubbling rate is held constant, provides a more direct evaluation of the effects of waste aluminum form (hydroxide vs. oxide), glass temperature, and glass composition on production rate. The changes in glass composition that were tested were based on the results from the glass formulation development studies (see Section 3.0) and were intended to demonstrate faster waste processing rates in comparison to previous DM100 tests with the same high-aluminum waste.

Figures 4.1.a - 4.1.d illustrate the glass production rates as moving hourly averages throughout the tests. Steady-state production rates for current and previous tests [2, 14, 15, 23-25, 44] conducted at the same constant bubbling rate at similar feed solids content are tabulated in Table 4.2; Figure 4.2 displays the results for tests conducted at a melt pool temperature of

1150°C. Taken together, the past and present test results show large differences in production rate with different waste/feed compositions and glass temperature. In previous tests [2], the effect of waste composition was greater than that of any other tested variable and this is further confirmed by the data from the current tests; in particular, the iron and chromium wastes were found to process at about twice the rate of aluminum limited wastes [2]. The higher processing rates are not only attributable to the differences in waste composition but also the lower waste loadings (24-28 vs. 45-47 wt. % oxide) used in many of the previous tests [14, 23]. Further, the current tests clearly demonstrate that feeds of the same composition but with the aluminum in the form of aluminum hydroxide instead of aluminum oxide process at faster rates. Comparison of tests conducted with both forms of aluminum using two different glass compositions (HLW-E-Al-27 and HWI-Al-16) show an approximate thirty percent production rate increase when aluminum oxide is replaced by aluminum hydroxide.

Importantly, the test results also showed that significant rate enhancements were achieved as a result of manipulation of the glass forming additive blends and glass compositions:

- For both forms of aluminum, production rates increased by ~30% by changing from the original HLW-E-Al-27 glass formulation to the new HWI-Al-16 glass formulation.
- A similar ~30% rate increase was also observed with the new HWI-Al-19 glass formulation for the aluminum hydroxide based waste simulant (the only one tested with that glass formulation). This formulation has a higher waste loading than the HWI-Al-16 formulation.
- The highest production rates (950 and 1500 kg/m<sup>2</sup>/day at 1150 and 1200°C, respectively) were obtained for the HWI-Al-19 glass formulation, which also had the highest waste loading tested (45 wt%).

These results corroborate the results from the glass formulation development studies, which indicated that higher melt rates could be achieved by manipulation of the blend of additives that are combined with waste. These rate enhancements are also consistent with previous DM100 tests, which showed a positive effect of increasing the boron concentration in the additives on the production rate [25].

As expected, glass production rates increased with increased glass pool temperature. Increases in response to the 50°C increase in melt pool temperature ranged from 47 to 86 percent in the current tests. The percentage melt rate increase per unit temperature increase of ~1%/°C is comparable to that obtained in previous work with WTP LAW feeds but, with the exception of the sodium and aluminum waste [2], somewhat larger than that obtained in previous DM100 tests with the Al-, Cr- and Bi-limited HLW simulants [2] (although it should be noted that a smaller temperature change (25°C) was used in those tests).

Overall, there were no significant difficulties in processing these feed and glass compositions during these tests. Cold cap conditions were similar to the range of conditions observed in previous tests with HLW feeds [14-17, 23-25, 44], particularly those feeds high in aluminum [2]. The feeds had a tendency to adhere to melter walls to form "shelves" and "bridges," some of which required manual dislodging from the walls. The feed used in Tests 7

and 8 formed a more fluid cold cap and formed softer deposits on the walls than the feeds used in the other tests. Interruptions during feeding totaled no more than about an hour for each test. These short interruptions were required in order to energize the top pair of electrodes, transfer feed to the feed tank, adjust the feed line in the peristaltic pump as a result of wear from the pump rollers, clear the feed tube, and, during Tests 1 and 6, to allow deposits in the plenum space to assimilate into the glass. There was no clear trend between feed composition (including aluminum source) and the need to clear the feed tube or dislodge deposits in the plenum. Spikes in feed rate often occurred immediately after feed transfers due to adjustments in tank mixer speeds and pump settings. During steady-state feeding periods, production rates typically varied by about ten percent from the mean rate. No foamy glass was observed in the glass discharge and no foam was observed on the melt pool surface or cold cap.

The results of various operational measurements that were made during these tests are given in Table 4.3. Glass temperatures are shown in Figures 4.3.a - 4.3.d, plenum temperatures in Figures 4.4.a - 4.4.d, electrode temperatures in Figures 4.5.a - 4.5.d, and glass resistance in Figures 4.6.a - 4.6.d; electrode power is included in the figures with electrode temperatures and glass resistance. The target bubbling rate of 9 lpm was maintained throughout the tests. Bulk glass temperatures (measured at 5 and 10 inches from the bottom of the melt pool) were largely within 10°C of the target glass temperatures of 1200°C and 1150°C throughout the vast majority of the tests. The test-segment-average bulk glass temperatures were 1190 - 1203°C and 1135 -1157°C for tests targeting glass temperatures of 1200°C and 1150°C, respectively. Glass temperatures closer to the top of the melt pool (measured at 16 and 27 inches from the bottom) are not reliable indicators of bulk glass temperatures as a result of their sensitivity to variations in the level of glass in the melter and gradients near the melt surface. As a result of the intentionally lower glass level, glass temperatures measured at these locations were even lower at the beginning of each test with a new feed composition, prior to the glass level in the melter being increased to above the upper pair of electrodes. Plenum temperatures typically ranged from 300 to 500°C, which is lower than the 550 to 650°C target. The lower temperature ranges were the consequence of maintaining a more complete cold cap. Similar plenum temperatures were measured in previous tests with aluminum limited wastes that also featured similar cold cap conditions [2].

The lower pair of electrodes was hotter than the upper pair of electrodes at the beginning of each test with a new feed composition due to the lower starting glass level. Once the melter was filled with glass above the top electrodes, the two electrode pairs are 0 to 150°C colder than the glass pool, depending on the measured points in the glass pool and electrodes. The bottom electrode, which was not powered, was 350 to 400°C colder than the powered side electrodes. Power supplied to the electrodes averaged between 18 and 26 kW for tests conducted at melt pool temperatures of 1150°C and 1200°C, respectively. The higher power requirement for the higher melt pool temperature also corresponds to higher production rates. The opposite trend is observed when power usage is normalized to glass production due to the amount of energy required to maintain the glass pool at the target melt temperature (i.e., the essentially constant idling power). The calculated glass pool resistance decreased dramatically as the melter was filled with glass, as would be expected. The melt pool resistance increased by about 0.015 to 0.02 ohms as the glass pool temperature was decreased from 1200°C to 1150°C.

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The gas temperature at the film cooler averaged between 275-287°C and depended on the plenum temperature, the amount of added film cooler air, and the temperature of the added film cooler air. Drops of less than seventeen degrees in gas temperature were observed across the (insulated) transition line; the high temperature is maintained in order to prevent condensation in the downstream filtration units.

Based on the results of the DM100 tests, the HWI-Al-19 glass formulation was selected for larger scale testing in the DM1200 HLW pilot melter. This formulation showed the highest productions rates, showed no processing issues in the DM100 test, meets all of the processing and product quality requirements, and has the highest waste loading.

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### SECTION 5.0 DM1200 OPERATIONS

Three tests with the high aluminum HLW simulant and the HWI-Al-19 glass composition were conducted between 8/6/08 and 8/16/08, producing almost 9 metric tons of glass. The total testing duration, including the time for water feeding and cold-cap burn-off, was 157 hours, during which over 24 metric tons of feed was processed. A summary of the test conditions and results is provided in Table 5.1. The tests were conducted to determine glass production rates for a high aluminum waste and optimized glass composition at two glass temperatures and bubbling rates. The HWI-Al-19 glass selected for the DM1200 tests had the highest processing rate of those evaluated on the DM100 (950 and 1500 kg/m<sup>2</sup>/day at 1150 and 1200°C, respectively) at the highest waste loading (45%). Each DM1200 test was nominally two days in duration, employed double-outlet prototypical bubblers, and processed feed with a solids content of 500 g glass per liter. The tests are summarized below in the order they were conducted:

- <u>Test 1</u>: Melt pool bubbling rate and glass temperature were optimized to yield the maximum production rate.
- <u>Test 2</u>: Glass temperature held constant at 1150°C and melt pool bubbling rate adjusted to maintain a production rate of 1050 kg/m<sup>2</sup>/day.
- <u>Test 3</u>: Glass temperature held constant at 1175°C and melt pool bubbling rate adjusted to maintain a production rate of  $1050 \text{ kg/m}^2/\text{day}$ .

The target glass production rate in Tests 2 and 3 was selected to provide direct comparisons with previous tests [6, 23].

### 5.1 Melter Operations

The DM1200 melter tests employed a prototypical ADS feed system, a single feed tube in the center of the melter, and a side-to-side electrode firing pattern. Two double ported bubblers (see Figure 1.5) positioned in a manner to mimic the WTP HLW melter (see Figure 1.6) were used throughout testing. In each test, the cold-cap-limited production rate was determined by visual observations of the cold cap and confirmed by the plenum temperature. The ADS feed system performed well in all tests. A summary of operational events is provided as Table 5.2. Operators observed the cold cap through view ports on the side of the melter on average every 20 minutes to guide melter operation, particularly the adjustment of bubbler air flow and feed rate. These observations are listed in Table 5.3.

Feeding was interrupted each day for about five minutes to collect feed samples. It was also paused once for eight minutes to remove a blockage in the film cooler, once for thirteen minutes to change a defective blower, and once for thirteen minutes to remove a blockage from the end of the feed tube. The majority of the film cooler deposits were readily removed by running a rod down through the film cooler and deposits on the feed tube were easily removed by tapping externally with a hammer. These feeding hiatuses did not compromise the objective of determining a steady production rate for any of the test segments. The feed used in these tests formed cold caps that often formed mounds, ridges, and cones under the feed nozzle. The aqueous feed was often observed boiling on the cold cap surface or running off onto the melt surface through holes in the cold cap generated from bubbling. A picture of one of these holes in the cold cap is provided in Figure 5.1. Ridges or mounds forming in the center of the melt pool often made direct observation of the melt surface opposite the view ports impossible. In these instances, the amount of cold cap was estimated by the amount of light visible as a result of openings in the cold cap. Ridges and mounds typically did not prevent free flow of the aqueous feed slurry across the cold cap surface and were managed by manipulating bubbler air flow. The configuration of the cold cap was constantly in flux throughout the tests; however, the extent of coverage was typically between 75 and 90 percent. The visual observations of the DM1200 melt pool is a key operational aspect of current DM1200 testing. In contrast to the operation of the LAW Pilot Melter, use of non-visual data, such as plenum temperature, have not been developed as reliable indicators of cold-cap conditions while processing HLW feeds [49]. In fact, high plenum temperatures can result from a high mound over a portion of the melt surface preventing feed from spreading across the melt surface and creating an opening on the glass surface. Without the visual evidence, an operator may conclude that feed rates should be increased, which could exacerbate the problem.

The glass production rates achieved during testing are provided in Table 5.1; these are illustrated in Figures 5.2.a and 5.2.b and are compared to rates achieved with other HLW feeds using the DM1200 with the optimized bubbling configuration in Table 5.4. The rate of 1500 kg/m<sup>2</sup>/day achieved during Test 1 was the highest production rate achieved with this configuration. This higher rate is due in part to higher feed solids content of 500g glass per liter as opposed to 340 to 430 g glass per liter used in previous tests. Coincidently, the rate of  $1500 \text{ kg/m}^2/\text{day}$  achieved on the DM1200 with optimized bubbling was the same rate obtained on the DM100 at 1200°C. Previous DM100 tests with HLW simulants have shown that production rates can be doubled while optimizing the bubbling as opposed to using the standard 9 lpm [2]. It is reasonable therefore that optimization of bubbling on the DM100 would have increased the rate of 950 kg/m<sup>2</sup>/day at 9 lpm to about 1500 kg/m<sup>2</sup>/day. The target glass production rate of 1050 kg/m<sup>2</sup>/day was easily achieved at glass processing temperatures of 1150 and 1175°C. This rate had been previously attained at 1150°C with HLW AZ-101 waste at 400 g glass per liter [6] with comparable bubbling (64 vs. 71 lpm), with HLW C-106/AY-102 waste at 340 g glass per liter [23] with more bubbling (90 vs. 71 lpm), and could not be obtained with HLW AZ-102 at 340 g glass per liter [23]. These results demonstrate that both waste composition and feed solids content have a big effect on production rate. Increasing the glass pool temperature by 25°C permitted the reduction of the bubbling rate by about forty eight percent (71 to 48 lpm) while maintaining a glass production rate of 1050 kg/m<sup>2</sup>/day. A reduction in bubbling of about thirty percent was observed with HLW AZ-101 waste at 400 g glass per liter with the same increase in glass temperature [6].

A variety of operational measurements recorded during these tests, including temperatures throughout the melter system, are given in Table 5.5. Data are collected and electronically logged every two minutes, and data and observations are also recorded manually

throughout the tests. The temperature for most of the glass pool was about 1150°C for the first two tests and about 30°C higher during the last test, as illustrated in Figures 5.3.a - 5.3.b. During the first test, the 1150°C melt temperature best suited the DM1200 power constraints while optimizing bubbling to achieve the maximum glass production rate. Conversely, the 1150°C glass temperature was targeted and successfully maintained during Test 2 while achieving the target glass production rate of 1050 kg/m<sup>2</sup>/day. The same targeted production rate was also maintained at the higher glass pool temperature of about 1180°C during Test 3. Glass temperatures near the surface of the glass pool (27" from the floor) were lower due to the thermocouples being in or near the cold cap. Aside from this excursion, bulk glass temperatures were relatively constant throughout the glass pool. Glass temperatures measured on the east side were 2 to 11°C higher than those on the west side of the melter. Plenum temperatures, given in Figures 5.4.a and 5.4.b, spanned a larger range during the testing, 300 to 900°C, than the 450 to 550°C target given in the Test Plan. The test segment and steady state average temperatures ranged from 482 to 714°C, which is much closer to this target range. Higher plenum temperatures were observed at the beginning of each test as the cold cap was being formed and as the thermowell and exposed thermocouple are exposed to openings in the cold cap. Conversely, lower plenum temperatures were observed when the thermowell or exposed thermocouple was encased in cold cap material. An example of the feed encasing a thermowell occurred during Test 3 at about 70 hours run time. Notice in Figure 5.4.b that the lower plenum temperature in the thermowell at 17" below the melter lid dropped to as low as 200°C due to cold cap material covering that portion of the thermowell, while locations higher on the thermowell and the exposed thermocouple showed measured temperatures of about 700°C due to openings in the cold cap.

The east and west side electrode temperatures were about 10 to 15°C below the glass temperatures on the respective side of the melter. These temperatures typically varied by no more than 20°C from the mean during each test, as shown in Figures 5.5.a - 5.5.b. The closeness of the side electrode to the glass temperature was a factor limiting the glass temperature used during Test 3. The bottom electrode, which was not powered during these tests, was about 60°C lower than the west electrode. The difference between the two side electrode temperatures was less during Test 1 due to the increased bubbling of the melt pool, which created better mixing. The discharge chamber and riser temperatures were largely maintained above 950°C throughout the tests. (The riser thermocouple is located about 4 inches above the bottom of the riser pipe, which is about 7.5 inches above the melter floor.) Gas temperature as a result of film-cooler and control-air dilution and depending on the position of the thermocouple in the plenum with respect to the cold cap. The film cooler was rinsed by a water spray every 12 hours during testing, resulting in a short-duration reduction to about 75°C in the film cooler outlet temperature.

Conditions in the glass pool are illustrated for electrical properties in Figures 5.6.a – 5.6.b, level and density in Figures 5.7.a – 5.7.b, and bubbling in Figures 5.8.a – 5.8.b. Power supplied to the electrodes was highest during Test 1, averaging 224 kV in response to higher bubbling and glass production rates. Power averaged 175 kW in both Tests 2 and 3 due to the glass production rate in both tests being 1050 kg/m<sup>2</sup>/day, even though the glass temperature was higher during Test 3 (1175 vs. 1150°C) and bubbling was higher during Test 2 (71 vs. 48 lpm).

Supplied power was relatively constant once the cold cap was established, varying by only about 10 kW from the average steady state value. The glass pool resistance decreased with bubbling over the course of these tests: Test 1 - 0.099 ohms at 124 lpm bubbling, Test 2 - 0.093 ohms at 71 lpm bubbling, and Test 3 - 0.083 ohms at 48 lpm bubbling. Conversely, average glass pool density at steady state conditions during testing decreased with bubbling: Test 1 - 2.22 g/cc at 124 lpm bubbling, Test 2 - 2.29 g/cc at 71 lpm bubbling, and Test 3 - 2.33 g/cc at 48 lpm bubbling. This decrease was observed even though the glass temperature increased, indicating that the effect of bubbling on glass density is greater than glass temperature. The resistance and density may also have been affected as the glass pool composition changed over to the target composition during testing. Glass pool depth varied between 30 and 34 inches in response to the continuous feeding of the melter and the periodic discharging of glass. Lance bubbling rates decreased over the course of testing, from an average high of 125 lpm while achieving maximum production rates during Test 1, to 48 lpm while maintaining 1050 kg/m<sup>2</sup>/day at an elevated glass pool temperature. Frequent changes to the bubbling rate as well as the distribution of bubbler air between the double ported lances were made to create an even cold cap across the melt pool surface.

## 5.2 Off-Gas System Performance

Tests on the DM1200 system at VSL have been used extensively to evaluate the performance of a pilot scale off-gas system that is prototypical of that designed for the WTP by BNI engineering [6, 16-23, 50]. In the current tests, data objectives primarily related to glass production rate and melter conditions required to achieve these production rates, not the performance of prototypical off gas system components. However, data for each of the off-gas system components and samples of solutions from the off gas system were collected and evaluated and are provided in this final report. Data are collected and electronically logged every two minutes and data and observations are also recorded manually throughout the tests. The average, minimum, and maximum values of the measured off-gas system parameters are given in Table 5.6. Target operational conditions for the system components such as sump temperatures, unit spray rates, and sump pH values that were not specified [3] were adapted from previous tests conducted on DM1200. The catalyst unit was bypassed in these tests. Plots of the typical sequence of gas temperatures through the DM1200 off-gas system at various locations are given in Figures 5.9.a, 5.9.b and 5.9.c for Test 1, Test 2, and Test 3, respectively. In summary, plenum gas from the melter is cooled by dilution with film cooler air to between 400 and 450°C, drops another 100°C by control air dilution and heat loss along the transition line, is guenched to 30 to 40°C in the SBS, and reheated to about 75°C to prevent condensation in the HEPA filtration unit. The exhaust is heated by another 5 to 10°C by the Paxton blowers, then quenched to 25 to 35°C in the PBS.

### 5.2.1 Melter Pressure

A vacuum on the melter of three to three and half inches of water was targeted and maintained throughout the tests. This is achieved by setting blower speeds and using a control air

system which constantly monitors the vacuum on the melter and injects sufficient air into the transition line immediately downstream of the film cooler to maintain a relatively constant vacuum on the melter. The computer-logged melter pressures measured at the instrument port and calculated control air flow rates for Test 1 and Tests 2 and 3 are plotted in Figures 5.10.a and 5.10.b, respectively. The range of control air flow rates of up to 80 scfm reflect the changes of melter exhaust volume in response to changes in the cold cap and feed rate, including pulsing of the feed (due to the ADS pump) throughout the tests.

The differential pressure across the film cooler and transition line is given in Figures 5.11.a and 5.11.b for Test 1 and Tests 2 and 3, respectively. The film cooler clogged and required manual cleaning once during Test 1 and once during Test 3. The film cooler was also rinsed every 12 hours with 5 liters of water. The typical film cooler and transition line differential pressures ranged between one and two and half inches water column and two and five inches water column, respectively. Measured values outside these ranges resulted from either manual or spray cleaning of the film cooler, blockage of the film cooler, particularly around 10 hours run time during Test 1, clogged sensor lines, opening of ports in the transition line and melter lid for sampling activities, and small pressure surges resulting from either the pulsed nature of feeding or changing conditions in the cold cap.

#### 5.2.2 SBS

The SBS quenches the melter exhaust, condenses much of the water from the melter feed, and removes the majority of the particulate in the exhaust stream. Many parameters of the SBS were recorded during testing, including inlet and outlet gas temperatures, pressures, and flow rates, pressure drops, sump temperature, heat exchanger inlet and outlet water temperatures, and flow rates. The amounts of heat removed by the SBS jacket, and the SBS inner cooling coil were calculated from the measured data, using the hourly averaged cooling water temperature increases (outlet temperature minus supply temperature) across the SBS inner cooling coil and cooling jacket multiplied by the same time-averaged water flow rate through each.

The SBS inlet and outlet gas temperatures are plotted in Figures 5.12.a and Figure 5.12.b for Test 1 and Tests 2 and 3, respectively. The average SBS inlet and outlet gas temperatures were 341°C and 46.2°C, 292°C and 39.3°C, and 303°C and 39.4°C during Test 1, Test 2, and Test 3, respectively. The higher temperatures during Test 1 were in response to the higher concentration of exhaust gases from feed resulting from the higher melter feed rate.

SBS inlet, outlet, and differential pressures are plotted in Figures 5.13.a and 5.13.b. During Test 1, the inlet gas pressure averaged -8.9 in. W.C., the outlet pressure averaged -41.2 in. W.C. and the pressure drop across the SBS averaged about 32.3 in. W.C. During Test 2, the inlet gas pressure averaged -7.9 in. W.C., the outlet pressure averaged -39.5 in. W.C. and the pressure drop across the SBS averaged about 31.6 in. W.C. During Test 3, the inlet gas pressure averaged -8.0 in. W.C., the outlet pressure averaged -40.2 in. W.C., and the pressure drop across the SBS averaged about 32.2 in. W.C.

The SBS down-comer annulus pressures are given in Figure 5.14.a and 5.14.b for Test 1 and Tests 2 and 3 respectively. The SBS off-gas temperatures in the down-comer measured at various depths (from 3 to 58 inches) and the SBS sump water temperature are given in Figures 5.15.a and 5.15.b, respectively. For Test 1 and Tests 2 and 3, the average SBS sump temperatures were 39.2°C, 33.1°C, and 33.0°C, which are 7.0°C, 6.2°C, and 6.4°C less than the SBS outlet gas temperatures, respectively. The measured off-gas temperatures decrease as the depth from the SBS lid increases due to cooling of the gas in the down-comer pipe by the surrounding SBS liquid.

Water temperatures at the SBS inner cooling coil inlet, inner cooling coil outlet/jacket inlet, and jacket outlet are given in Figures 5.16.a and 5.16.b for Test 1 and Tests 2 and 3, respectively. The average water temperature difference was 15.5°C across the SBS inner cooling coil and 3.9°C across the jacket during Test 1. The average water temperature difference was 12.4 °C across the SBS inner cooling coil and 2.9°C across the jacket during Test 2. The average water temperature difference was 12.0°C across the SBS inner cooling coil and 3.2°C across the jacket during Test 3.

The SBS cooling coil/SBS jacket water flow rates are plotted in Figures 5.17.a and 5.17.b and averaged 28.7 gal/min, 27.0 gal/min, and 27.8 gal/min for Tests 1, 2, and 3, respectively. The corresponding amounts of heat removed by the SBS inner coil and jacket are shown in Figures 5.18.a and 5.18.b. For Test 1, heat removal averaged 95.9 kW by the SBS inner cooling coil and 15.5 kW by the cooling jacket. This corresponds to about 86% of the heat load to the SBS being removed by the inner cooling coil and 16.6 kW by the cooling jacket. For Test 2, heat removal averaged 72.0 kW by the SBS inner cooling coil and 16.6 kW by the cooling jacket. This corresponds to about 81% of the heat load to the SBS being removed by the cooling jacket. For Test 3, heat removal averaged 71.5 kW by the SBS inner cooling coil and 19.2 kW by the cooling jacket. This corresponds to about 82% of the heat load to the SBS being removed by the SBS being removed by the inner cooling coil and 19.2 kW by the cooling jacket.

One of the functions of the SBS is to condense water that originated in the waste feed. Figures 5.19.a and 5.19.b compare the amount of water fed to the total volumetric accumulations in the SBS over the course of each test. The amount of water fed into the melter is proportional to the amount of water in the feed and the slurry feed rate; hence, Tests 2 and 3 which targeted the same feed rate showed constant water feed rate to the melter. The difference between the amounts of SBS water coming from the feed and the amounts blown down from the SBS sump represents the amount of water carried out in the off-gas stream as a result of it being saturated at the SBS sump temperature, as well as a small amount of entrained droplets. This amount is largely determined by the SBS sump water temperature, which was about 6°C higher during Test 1 than in Tests 2 and 3. As a result this temperature difference, the total volume of water exiting the SBS was about the same for Test 1 and the combined volume for Tests 2 and 3, despite the difference in run time.

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#### 5.2.3 WESP

The WESP is used primarily to remove fine, often water soluble particles from the exhaust stream that are not efficiently removed by the SBS. The inlet and outlet gas temperatures, differential pressure across the WESP, and the WESP current and voltage were measured and recorded by the computer data acquisition system. The WESP inlet and outlet gas temperatures are plotted in Figures 5.20.a and 5.20.b for Test 1 and Tests 2 and 3, respectively. A one to two degree increase in temperature is observed in the exhaust temperature as gas passes through the WESP. The periodic downward spikes in the WESP outlet temperature are a result of the daily deluge of the WESP to wash collected deposits off the electrodes and into the WESP sump. The WESP differential pressures and outlet gas flow rates are plotted in Figures 5.21.a and 5.21.b for Test 1, and Tests 2 and 3, respectively. A pressure drop of about two and a half inches across the WESP was observed during testing. The typical wet gas flow rate exiting the WESP was about 230 scfm.

The amount of liquid accumulated in the WESP is plotted as a function of run time in Figure 5.22.a and Figure 5.22.b, where it is compared with the amount of fresh water sprayed into the WESP during Test 1 and Tests 2 and 3, respectively. The inlet spray water was targeted at 2.0  $\pm$  0.2 gph; however, the actual spray water flow rate was  $\approx$  1.7 gph because of the limitations of the spray nozzle. As evident from both figures, spray water accounts for the majority of the liquid accumulation in the WESP. The difference between accumulated liquid and fresh water sprayed is equal to the amount of liquid removed from the off-gas, which is also plotted in Figure 5.22.a and Figure 5.22.b. The WESP electrodes were deluged daily, as planned, with water at a nominal rate of 12 gpm for 3.33 minutes.

The WESP voltage and current are plotted as functions of run time in Figure 5.23.a and Figure 5.23.b for Test 1 and Tests 2 and 3, respectively. The current was set at 17 mA with voltage not to exceed 32 kV prior to introducing water into the melter. The current was subsequently adjusted to that target during testing to the extent possible. During Test 1, the target current was approximated but it could not be reached during Tests 2 and 3 within the 32 kV voltage limit. The power stabilized shortly after the daily deluges.

#### 5.2.4 Secondary Off-gas System

A HEME filtration unit (HEME 1) follows the WESP in the off-gas system to remove water droplets that may be present in water-saturated gas exiting the WESP. For Test 1, and Tests 2 and 3, the outlet gas temperature and differential pressure are plotted in Figure 5.24.a and 5.24.b. The typical pressure drop across HEME 1 during testing was about 1.5 in W.C.

The HEME is followed in the off-gas system by a heater, a HEPA filter (HEPA 1), and a Paxton blower (Blower 1). The purpose of the heater is to ensure that water-saturated gas exiting HEME 1 is heated above its dew point before passing through the HEPA filter in order to prevent moisture condensation in the HEPA filter. The outlet gas temperature and the pressure differential across the HEPA filter are the two parameters monitored by the off-gas data acquisition system; for Test 1 and Tests 2 and 3, these are shown in Figure 5.25.a and 5.25.b,

respectively. The typical pressure drop across the HEPA filter was about half an inch water column throughout testing.

A vacuum is maintained on the melter by a pair of redundant Paxton blowers (Blowers 701 and 702) immediately downstream of the HEPA filtration unit and a blower (Blower 801) downstream of the packed bed scrubber. During Test 2 at 23.1 hours, Blower 701 failed and Blower 702 was placed in service while the head on Blower 701 was replaced. Also, the speed of the redundant blowers was increased and the speed of Blower 801 was increased to reduce the temperatures in the redundant blowers.

Downstream of the HEPA filter and Paxton blowers in the off-gas train is the packed bed caustic scrubber (PBS) to remove iodine and acid gases from the off-gas stream. The PBS sump solution is derived from process water; caustic solution (25% NaOH) is added to control the solids content and pH of the scrubber liquid. The PBS inlet gas temperature, pressure drop across, sump temperature, and sump pH for Test 1 and Tests 2 and 3 are shown in Figures 5.26.a and 5.26.b, respectively. The typical pressure drop across the PBS during testing was two and a half inches water column. The inlet gas at about 80-85°C was quenched to about 25-30°C in the PBS during testing. The saw tooth appearance of the measured pH values around 9 and 10 results from the automated additions of NaOH solution to maintain the relatively constant pH.

#### 5.2.5 SBS, WESP, HEME, and PBS Process Fluids

One-liter samples were collected from the SBS, WESP, PBS, and HEME sumps at the end of testing. Samples were subjected to total dissolved solids (TDS) and total suspended solids (TSS) determinations by gravimetric analysis of filtered material and the evaporated filtrate. An additional sample was filtered to generate solids and filtrate for complete chemical analysis, which included pH determination, direct current plasma atomic emission spectroscopy (DCP-AES) analysis for metals, and ion chromatography for anions; the dried filtered solids underwent microwave-assisted acid dissolution prior to chemical analysis. The terminal SBS, WESP, PBS, and HEME sump samples as well as the total volume of solution removed from each sump are listed in Table 5.7; the first letter in the sample name is "S", "W", "H", and "P" for the SBS, WESP HEME, and PBS samples, respectively. The "A" and "B" suffixes in the name of the WESP samples correspond to samplings immediately before and after the 40-gallon deluge. The "1" or "2" after the "H" in the HEME sample name correspond to the first HEME located downstream of the WESP and the second HEME located downstream of the PBS, respectively. The analyzed chemical compositions for samples taken at or near the end of each test are provided in Table 5.8.

The amount of solution removed from the first HEME (immediately downstream of the WESP) at the end of each test and corresponding chemical analysis is given in Table 5.8. The HEME was continuously sprayed at  $\sim 0.2$  gal/hr, resulting in the addition of about 10 gallons of water for each of the three nominally 50 hour tests. The liquid volume accumulated during each of the tests was five to ten gallons greater than the amount sprayed as a result condensation. The PBS was blown down as required to maintain constant volume.

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The pH of sump solutions is affected by constituents removed from the exhaust by each component and the amount of process water or sodium hydroxide added. Feed processed in the current tests contains relatively higher ratios of halides, sulfate, nitrate, and boron to alkali metals, reductants, and silica, as compared to other HLW feeds previously processed [16-23]. As a result, the pH in the primary off-gas system effluents were significantly more acidic than in previous tests on DM1200 with HLW feeds. The pH values of less than 2 are more similar to values measured while processing some LAW wastes than to those for HLW wastes, which typically result in near-neutral solution pH values. The pH of the post deluge WESP solution is closer to neutral due to dilution with the deluge tap water than is the solution sampled from the WESP prior to the deluge. The pH of the solution from the HEME that follows the WESP is very similar to that of the WESP solution, whereas the pH of the solution from the HEME that follows the result around 9 during testing by the addition of 25% sodium hydroxide solution, which is reflected by measured pH values near 9 in the PBS solutions.

The chemical composition as well as the distribution between dissolved and suspended solids fractions was measured on sump solutions from the end of testing and is provided in Table 5.8. Most of the chemical species measured in off-gas system solutions were dissolved due to the efficiency of the SBS at removing coarse particles, which are often insoluble, and the acidic conditions in the primary off-gas system components. As expected, the most abundant solids in the SBS solution were soluble species such as halogens, boron, alkali metals, nitrate, and sulfate. These species are readily volatilized from the glass and cold cap in the melter as soluble salts. Similar results were obtained from analysis of SBS solutions in tests with other HLW simulants [6, 16-23]. Dissolved chlorine and fluorine were observed in significant proportions in all of the tests, even though chlorine was not targeted in the feed, indicating that these halides were present in the feed as a contaminant. In contrast to previous tests with HLW simulants [6, 16-23], aluminum was also abundant in these solutions as a dissolved species due to the relatively high concentration in the feed and the acidic conditions in the sump solution. The WESP solutions contain mainly volatile salts (alkali halides, borates, and sulfates) carried over from the SBS, and contaminants from previous tests. Impurities in the feed and tap water are major constituents in the WESP solutions. The concentration of suspended solids in the WESP post-deluge sample is over an order of magnitude higher than for pre-deluge solutions; this material was presumably material that was dislodged from the electrodes. Conversely, the dissolved solids concentration in the post-deluge sample is about a third of the pre-deluge sample due to dilution by the deluge water. The chemical analysis of solution from the first HEME indicates that the solutions are essentially diluted WESP solutions, which is consistent with the HEME collecting mist carried over from the WESP. Nitrate and nitrite are higher in the HEME solutions than the contemporaneous WESP solutions, suggesting that the HEME is more efficient at removing these constituents from the exhaust stream. The relatively high concentrations of sodium, halides, nitrogen oxides, and sulfate in the PBS indicate that the PBS is functioning as intended. The chemical analysis of solution from the second HEME indicates that the solutions are essentially diluted PBS solutions, which is consistent with the HEME collecting mist carried over from the PBS. Since many of the components include tap water sprays or are initially charged with tap water, elements common in tap water, such as calcium, are over represented in the solutions as compared to the amounts in the target feed composition.

### SECTION 6.0 FEED SAMPLE AND GLASS PRODUCT ANALYSIS

### 6.1 Analysis of Feed Samples

#### 6.1.1 General Properties

Feed samples from each test were analyzed to confirm physical properties and chemical composition. Samples were taken during melter testing from an inline sampling port. Sample names, sampling dates, and measured properties are given in Table 6.1. Density and pH were measured on all samples; at least one sample per unique waste composition from the DM100 tests and all DM1200 samples were analyzed for water content, glass conversion ratio, rheological properties, and oxide composition by XRF. The measured glass conversion ratios for all feed samples except one were within ten percent of the target on a weight per weight basis, validating the use of the target conversion ratio for calculating glass production rates. Samples from the DM100 tests were closer to the target values due in part to more water being used to flush feed lines during transfers to the DM1200 feed tanks. The water content, density, glass yield, and pH varied within a narrow range except for the pH values for DM100 Tests 1 and 2. The higher pH in these samples is attributable to the boron source; borax was the source of boron in the first two DM100 tests whereas boric acid was the boron source in all other tests. Changing of the aluminum source from oxide to hydroxide increased the pH by only a quarter of a unit.

#### 6.1.2 Rheology

Samples of the melter feeds that were used for these tests were also subjected to rheological characterization. The results from rheological characterization of a variety of other melter feeds and waste simulants, as well as the effects of a range of test variables, are described in detail in a separate report [51]. Melter feeds were characterized using a Haake RS75 rheometer, which was equipped with either a Z40DIN or a FL22-SZ40 sensor. A typical set of measurements consists of identifying the flow characteristics of the slurry by measuring the shear stress on the slurry at controlled shear rates and temperatures. In these measurements, the shear rate values are preset and are increased stepwise from 0.01 s<sup>-1</sup> to 200 s<sup>-1</sup> (70 s<sup>-1</sup> for FL22-SZ40) with a sufficient delay (typically 15 to 30 seconds) between steps to ensure that the shear stress is allowed to fully relax and therefore is measured at equilibrium. This approach is somewhat different from the "flow curve" approach in which the shear rate is ramped up to some maximum value and then ramped back down to produce a hysteresis curve that is dependent on the selected ramp rate. The viscosity of the sample as a function of the shear rate is then calculated as the ratio of the shear stress to the shear rate. The yield stress data for the melter feeds were measured using a controlled-stress mode in which the torque on the rotor was slowly increased while the resulting deformation of the fluid was monitored. The discontinuity in the measured deformation-torque curve was identified as the yield stress. It should be noted that this direct measurement of the yield stress can be quite different from the value that is often reported

as the yield stress, which is obtained by extrapolation of the shear stress-shear rate curve to zero shear rate. All of the measurements in this work were made at 25°C; previous work [51], which examined a range of temperatures, showed a relatively weak effect of temperature.

Measured values for viscosity at selected shear rates and the yield stress are shown in Table 6.2. The data indicate that the change of the form of aluminum in the waste and the changes in the additive blends do not have substantial effects on the rheological properties of the feed samples. A small increase in feed viscosity and yield stress may be attributable to changing the boron source from borax to boric acid. All feeds were processed by the feed system without significant difficulties.

### 6.1.3 Chemical Composition

The methods used for analysis of feed sample chemical compositions are described in Section 1.5. The boron, fluorine, and lithium oxide target values were used for normalizing the XRF data since their concentrations were not determined by XRF. These results, compared to the target composition in Table 6.3, generally corroborate the consistency of the feed compositions and show good agreement with the target compositions for the major elements. All oxides with target concentrations greater than one percent deviated by less than 10% from target for feed samples from DM1200 tests. Similarly, most oxides in feed samples from DM100 tests also passed this criterion; however, phosphorus and bismuth deviated by as much as sixteen percent from target. However, the absolute deviation for bismuth and phosphorus oxides was always less than 0.18 wt%. The iron oxide concentration in the feed sample from DM100 test 5 was about nineteen percent above the target value; however, glass made from this feed during melter testing had measured iron concentrations within ten percent of the target value. This deviation may be attributable to contamination during sampling or crucible melting. The composition of this feed is further corroborated by comparison to the product glasses (see Section 6.2.1), which shows all oxides with concentrations greater than 1 wt% in the target composition to be within about 10% of the target, except for phosphorus during some tests.

Low concentrations of manganese, neodymium, and strontium in some feed compositions, were measured, even though they are not included in the target composition. Also, common elements such as magnesium, titanium, zirconium, potassium, barium, and chromium, when targeted at low concentrations, were typically above these targets. These positive deviations are often observed in melter feeds due to their ubiquity in the raw materials used to make up the simulants and in the glass forming additives. Analysis of the product glass using a method other than XRF (see Section 6.2.1) indicates that cadmium is present at the low target concentrations; however, these concentrations are below the sensitivity of the XRF for these glass matrices. Analyzed sulfur concentrations are below target concentrations due to volatilization during sample preparation. Lead, which was targeted at low levels in the glass product (0.40-0.41 wt%), was measured at about ninety percent of target in the feed samples. None of these small deviations would affect the determination of glass processing rates or negatively impact product quality.

### 6.2 Analysis of Glass Samples

In the DM100 and DM1200 tests, almost two metric tons and nine metric tons of glass were produced, respectively. The glass was discharged from DM100 periodically into 5-gallon carbon steel pails and from DM1200 into 55 gallon drums, using airlift systems in both cases. The discharged product glass was sampled at the end of each test by removing sufficient glass from the top of the cans for total inorganic analysis. Product glass masses, discharge date, and analysis performed are given in Tables 6.4 and 6.5. Glass samples were also obtained by dipping a rod into the glass pool at the beginning and end of each test. These "dip samples" underwent visual examinations to detect the presence of separate sulfate or crystalline phases on the glass surface.

### 6.2.1 Compositional Analysis of Discharge Glasses

All discharge glass samples were crushed and analyzed directly by XRF. The target values for boron and lithium oxides, which are not determined by XRF, were used for normalizing the XRF data to 100 wt%. Fluorine analysis by XRF required a polished monolith as opposed to the standardized ground glass preparation used for the other elements. Approximately, every fifth discharged glass sample was directly analyzed for fluorine; fluorine concentrations of other glasses were interpolated in between the measured values. The XRF analyzed compositions of discharged glass samples are provided in Tables 6.6 and 6.7. The majority of the XRF analysis results compared very favorably to their corresponding target values and feed sample analyses (see Section 6.1.3). Oxides with a target concentration greater than one weight percent showed around, or below, 10% deviation from the target values. The only exception is phosphorus oxide which had an absolute deviation of no more than 0.15 wt%. Similar to the feed sample analyses, common elements such as magnesium, titanium, zirconium, and potassium targeted at low concentrations were above their respective targets. Several elements were present in the melt pools prior to these tests but were not included in the current target compositions. Most notable was strontium oxide at about four weight percent in the first glass discharged from the DM100 tests; other elements at much lower concentrations were cerium and neodymium in the DM100 melt pool, and lanthanum, neodymium, antimony, and strontium in the DM1200 melt pool. Sulfur and fluorine are below target for almost all glasses due to volatilization from the glass pool and cold cap.

Corroborative analysis using DCP on solutions of acid-dissolved glass was performed on select glasses produced from each test; the results are compared to the XRF analysis in Table 6.8. Values for all the major oxides compare favorably with the XRF analysis and target composition except for sodium, which often exhibits a low bias using this procedure [19, 52]. Low biases for aluminum and chromium using the DCP method were observed in previous tests with high aluminum concentrations [2]. It is possible that the aluminum concentrations in these glasses may be higher than the sample preparation method can dissolve or keep in solution with the acid mixture currently being used. Consequently, the XRF values are to be preferred. The closeness of the DCP boron and lithium analyses to the target (deviations less than 10%) validates the use of the target boron and lithium concentrations for normalizing the XRF data.

Compositional trends for selected constituents shown in Figures 6.1.a-6.1.c for DM100 tests and 6.2.a-6.2.c for DM1200 tests illustrate the closeness to targets at the end of tests with each composition. Exceptions include volatile species such as sulfur and fluorine, which remain significantly below their target concentrations as a result of significant release to the melter exhaust. Fluorine concentrations also decreased during idling periods in between DM100 tests. At the onset of testing, aluminum, calcium, and fluorine increase in concentration at the expense of silicon, iron, zirconium, and zinc as the steady-state composition is approached. Although not depicted in the figures, oxides present in the melt pool at the beginning of testing but not present in the target glass composition decrease in concentration to the point of being non-detectable by the end of testing. The concentrations of most oxides do not change significantly from composition to composition during DM100 testing since the waste composition is constant, the waste loading changes are very small, and significant additive changes occur for only boron and silica. Complete compositional turnover of the DM1200 melt requires the majority of the test duration due to the approximate two metric ton capacity of the melt pool; therefore the melt pool composition is evolving toward the target composition over nearly the entire DM1200 testing period.

### 6.2.2 Chemical Durability of Discharge Glasses

Glass discharge samples from the end of each DM100 test were evaluated for chemical durability using the PCT and TCLP methods. The PCT results are compared to those for the benchmark DWPF-EA glass in Table 6.10 and the TCLP results are compared to the WTP delisting limits [53, 54] and Universal Treatment Standard (UTS) limits in Table 6.11. The chemical durability determined by both of these methods is excellent for the melter glasses. All measured PCT concentrations and normalized leach rates on discharge glass samples are at least one hundred forty times lower than the corresponding values for the DWPF-EA glass. All regulated TCLP leachate concentrations are less than 1 mg/l and more than a factor of nine less than WTP delisting limits. All measured concentrations are also well below the UTS limits. The chemical durability of these glasses is similar to that for those previously produced from aluminum limited waste and better than that for those produced from aluminum plus sodium limited waste [2]. These results confirm that glasses can be formulated from a variety of waste loading limiting constituents without compromising the quality of the vitrified product.

#### 6.3 Glass Pool Samples

A list of all dip samples including sample names, sampling dates, target glass pool temperature, and visual observations of secondary phases is provided in Table 6.11. These samples were used to detect the presence of any secondary phases on the melt pool surface. There was no visual evidence of secondary phases in any of the samples as either material adhering to the sampling rod indicative of a surface layer or macroscopic features indicative of crystalline phases in the glass.

#### 6.4 Cold Cap Samples

Samples from DM100 Test 8 and DM1200 Test 1 were taken from the cold cap using a "spoon" device that was introduced into the melter from the top. The spoon penetrated the cold cap and was lowered into the melt, then withdrawn in such a way that it captured a piece of the broken cold cap while allowing the glass melt to drain as much as possible. The sample is then withdrawn from the melter, allowed to cool to room temperature, removed from the sampling device and prepared for SEM-EDS analysis. The samples obtained in this manner are typically a composite of desiccated melter feed, glass, and transitional phases in between these two extremes. Examination of the transitional phases can provide insight into the slower melting components of the melter feed.

### 6.4.1 DM100 Cold Cap Sample

The DM100 cold-cap was collected and split into two samples labeled BLZ-O-35A and BLZ-O-37A. Figure 6.3.a shows the majority of undigested particles, including quartz and calcium silicate. These particles are embedded in the fine mixture phase that is high in Al, Si, Ca, and Fe. Regions in dark contrast are a glass that is rich in light elements such as sodium, boron, lithium, aluminum, and silicon.

Figure 6.3.b shows a region close to the glass phase in the sample. At this location, at least 85% of the melter feed was converted to glass. Undigested particles include quartz and hematite in two forms: tiny needles of hematite aggregated to 20 - 100  $\mu$ m in size and individual needles of about 10-30  $\mu$ m in length. These hematite crystals usually contain nickel, chromium, or titanium. Crystals of Fe-Cr-Ni spinel are present as larger 5 to 20  $\mu$ m crystals as well as smaller 1-2  $\mu$ m crystallites associated with hematite. Other crystals include zirconia-rich phases, presumably zircon, and chromium oxide.

#### 6.4.2 DM1200 Cold Cap Sample

The DM1200 sample was identified as O-12J-106A. Figure 6.4 shows undigested particles of  $Al_2O_3$ ,  $ZrO_2$ ,  $Fe_2O_3$  and  $CaSiO_3$ . Undigested particles rich in  $Al_2O_3$  are the most abundant. These particles are cemented together by a fine mixture phase with small bright particles rich in iron, chromium, and zirconium.

### SECTION 7.0 MONITORED OFF-GAS EMISSIONS

#### 7.1 Particulate Sampling

The melter exhaust was sampled for metals/particles according to 40-CFR-60 Methods 3, 5, and 29 at steady-state operating conditions during each test segment. The concentrations of off-gas species that are present as particulates and gaseous species that are collected in impinger solutions were derived from laboratory data on solutions extracted from air samples (filters and various solutions) together with measurements of the volume of air sampled. Particulate collection required isokinetic sampling, which entails removing gas from the exhaust at the same velocity that the air is flowing in the duct (40-CFR-60, Methods 1-5). Typically, a sample size of 30 dscf was taken at a rate of between 0.5 and 0.75 dscfm. Total particulate loading was determined by combining gravimetric analysis of the standard particle filter and chemical analysis of probe rinse solutions. An additional impinger containing 2 N NaOH was added to the sampling train to ensure complete scrubbing of all acid gases and, particularly, iodine. The collected materials were analyzed using direct current plasma atomic emission spectroscopy for the majority of the constituents and ion chromatography (IC) for anions. Melter emission fluxes are compared to feed fluxes in Table 7.1 for DM100 tests and Table 7.2 for DM1200 tests. Notice the distinction that is made between constituents sampled as particles and as "gas". The "gaseous" constituents are operationally defined as those species that are scrubbed in the impinger solutions after the air stream has passed through a 0.3 µm heated filter. All samples but one are well within the 90 - 110% limits for isokinetic sampling. The particle filter during the last sample occluded shortly after the start of sampling, which limited the duration and resulted in a less than isokinetic sample.

Particulate emissions from both melters constituted 0.04 to 0.29 percent of feed solids. This level of carry-over is less than that measured for previous tests with chromium, bismuth, aluminum, and aluminum plus sodium limited HLW wastes (0.06 - 0.57) [2], HLW AZ-102 (0.57 - 1.47 percent) [44], and HLW C-106/AY-102 SIPP (0.61 to 0.81 percent) [15] simulants processed on the same melter. The higher carry-over in many of the previous tests is due to higher proportions of volatile species in the feed such as rhenium, cesium, and halogens. The carry-over is lower but still comparable to previous tests conducted with HLW AZ-101 simulants while bubbling the melt pool [14]. In all paired DM100 tests with the same feed, particulate emissions were higher at the higher glass temperature of 1200°C than at 1150°C. An increase in particulate emissions was also observed on the DM1200 with a glass pool temperature increase of only 25°C and, separately, with an increase in melt pool bubbling. These trends are in contrast to the lack of clear emissions trends as a function of glass temperature and bubbling in the earlier DM100 tests with Al, Al+Na, Bi, and Cr limited HLW streams [2] but in keeping with numerous studies using HLW streams that show increased particulate carry-over with increased melt pool bubbling [6, 16-21] and previous tests showing increased particulate melter emissions with increasing glass processing temperature [14, 55-58]. It is possible that on the DM100 the effect of the smaller 25°C increase in temperature in the earlier tests [2] may be within the variability

of the measurements [1, 58] whereas the 50°C increase in the current tests was sufficient to observe increased emissions. No trend of increased particulate emissions with increasing boron concentration was observed despite the somewhat volatile nature of boron; larger increases in feed boron concentration may be required to see this effect. The amount of moisture collected in each sampling train was proportional to the feed rate; therefore, in the DM100 paired tests, the amount of moisture collected in the tests conducted at a 1200°C melt temperature was always higher than moisture collected in tests conducted at 1150°C.

As expected, the feed elements emitted at the lowest melter DF were clearly fluorine and sulfur. Another element exhibiting some volatile behavior was boron. The relative volatility of barium, cadmium, and titanium are difficult to evaluate due to the low target concentrations in the feed. Emissions of chlorine were measured during all tests even though they were not included in simulant recipes or in the analyzed compositions. Chlorine has frequently been observed in melter emissions when not present in feed recipe or at levels exceeding the amount in the feed recipe due to its ubiquity in raw materials, presence in tap water, and high volatility at glass melting temperatures. Boron, sulfur, and the halides were the only elements detected in the impinger solutions collected downstream of the heated particle filter in the sampling train, which constitutes the "gas" fraction of the melter emissions.

## 7.2 Gases Monitored by FTIR

Melter emissions were monitored in each test for a variety of gaseous components, most notably CO and nitrogen species, by Fourier Transform Infra-Red Spectroscopy (FTIR). The off-gas system temperature is maintained well above 100°C beyond the sampling port downstream of the DM100 HEPA filter to prevent analyte loss due to condensation prior to monitoring. The data, therefore, represent the relative concentrations of volatile gaseous species in the melter exhaust. During the DM1200 tests, the exhaust stream was sampled at the outlets of several prototypical components (melter, SBS, WESP, and PBS) to discern the effect these components have on the volatiles in the exhaust stream. It should be noted, however, that the off-gas system component most responsible for the removal of nitrogen oxide and volatile organics, the TCO-SCR catalyst unit, was bypassed in these tests due to the relatively low concentrations of these components in the exhaust stream. Also, a single FTIR unit was used for all of the measurements and, therefore, locations were sampled sequentially and not simultaneously.

A summary of the range and average concentrations of gaseous species monitored during the DM100 tests is provided in Table 7.3; similar information for the DM1200 tests is provided in Tables 7.4 - 7.6. The concentrations of two of the monitored species are plotted in Figures 7.1 - 7.8. The analytes listed in Tables 7.3 - 7.6 are those that were thought likely to be observed during the tests based on previous work; no other species were detected in the off-gas stream by FTIR. Generally, emissions from the DM100 were relatively low as a result of the low concentrations of nitrogen, organic carbon, ammonia, and halogens in the feed. The most abundant nitrogen species monitored was NO, which is in keeping with previous melter tests with both HLW and LAW feeds. No nitrogen was detected as other species, except for NO<sub>2</sub>, which was 10 to 20 times lower in concentration than NO. Similar trends were observed in melter emissions during the DM1200 tests with a greater proportion of nitrogen oxides as NO<sub>2</sub> (approximately 6:1, NO:NO<sub>2</sub>) and very low concentrations of N<sub>2</sub>O, nitric acid, and nitrous acid in some of the tests. The variability in the NO concentrations shown in Figures 7.1 - 7.4 is attributable to the dynamic conditions in the cold cap and is in keeping with previous melter tests. The concentration of water in the melter exhaust increased with increasing feed rate in both the DM100 and DM1200 tests. Consistent with the gaseous fluorine concentrations observed using the Method 5-type sampling (see Section 7.1), HF was observed throughout the testing by FTIR. The variations in emissions over the course of each test segment are due in part to changes in the melt pool cold cap. Hydrogen fluoride concentrations were lower at the beginning of testing due to the lack of fluorine in the glass pool and the processing time required for the glass to reach steady-state concentration with respect to fluorine. Also consistent with the Method 5type results, low sulfur dioxide concentrations were monitored in each test; however, gaseous sulfur emissions can also be present in forms other than sulfur dioxide that are not monitored by the FTIR, such as sulfuric acid. Measured concentrations for most constituents through the DM1200 exhaust system were very similar. This confirms the expectation that the SBS, WESP, HEME, and PBS do not remove significant proportions of nitrogen and carbon oxides. Conversely, moisture and some minor components such as HF and SO<sub>2</sub> are greatly reduced in concentration by removal in the SBS. These trends can be readily discerned by comparing NO concentrations in Figures 7.5 and 7.7, which do not show significant differences at the various sampling localities, and HF concentrations in Figures 7.6 and 7.8, which are noticeably higher in the melter exhuast. Another aspect of the NO emissions is the high degree of variation during testing throughout the off-gas system, as can be discerned from the concentration ranges.

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## SECTION 8.0 SUMMARY AND CONCLUSIONS

A series of tests was conducted to develop HLW glass compositions for the aluminum limited waste composition specified by ORP that maximize processing rates while maintaining high waste loadings and acceptable glass properties. This was accomplished through a combination of crucible-scale tests, confirmation tests on the DM100 melter system, and demonstration at pilot melter scale (DM1200). New glass formulations and additive blends were developed with the objective of increasing processing rates while maintaining high waste loadings. The formulations used for melter testing were selected based on the test results from Vertical Gradient Furnace tests, DM10 Feed Consumption tests, and a series of crucible melt glasses that were prepared and characterized. Each of the selected formulations meets all of the product quality, processability, and waste loading requirements. The DM100 melter tests evaluated the effects of a 50°C increase in glass processing temperature, aluminum waste form  $(Al_2O_3 vs. Al(OH)_3)$ , and optimized additive blends on glass production rates. This same melter was used in previous tests with aluminum limited and other HLW wastes thus providing comparisons between the previously collected data on melter processing characteristics, partitioning to off-gas, and formation of secondary phases. Once DM100 tests were completed, one of the compositions was selected for further testing on the DM1200; the DM1200 system has been used for processing a variety of simulated WTP waste streams. Tests on the larger melter provide processing data at one third of the scale of the actual WTP HLW melter and therefore provide a more accurate and reliable assessment of production rates and potential processing issues.

For each of the four selected feed formulations, DM100 melter testing was performed in two 50-hour test segments. These test segments employed glass pool temperatures of 1150°C and 1200°C, bubbling rates of 9 lpm, and 500 g glass per liter feed solids concentration. Over three and a half metric tons of feed was processed to produce almost two metric tons of glass. Analysis was performed on discharge and glass pool samples throughout the tests for total composition. All of the melter tests were successfully completed with no evidence of processing issues. Glass production rates with the aluminum limited waste at a glass pool temperature of 1150°C were increased from 550 kg/m<sup>2</sup>/day in the previous tests to 950 kg/m<sup>2</sup>/day as a result of changes to the feed composition while maintaining the same high waste loading (45 wt%). Glass production rates as high as 1500 kg/m<sup>2</sup>/day were achieved at the glass pool temperature of 1200°C. Data collected during DM100 tests also demonstrated the effect of the aluminum source on production rates, with significantly higher rates observed with aluminum hydroxide than with aluminum oxide. Tests with the aluminum limited waste on the DM1200 demonstrated glass production rates of 1500 kg/m<sup>2</sup>/day at 1150°C melt pool temperature with optimized bubbling. The target production rate of 1050 kg/m<sup>2</sup>/day was easily obtained with a modest amount of bubbling and the nominal glass pool temperature of 1150°C as well as with an even lower bubbling rate at a modestly higher glass pool temperature of 1175°C. No significant processing problems were encountered on the DM1200 while producing almost 9 metric tons of glass from over 24 metric tons of feed.

During each test, melter exhaust was sampled for particulate and gaseous species to determine the effect of the variations in feed composition and glass temperature on emissions. Total particulate carry-over from both melters into the off-gas stream was only 0.04 to 0.29 percent, which is below the range measured previously on the DM100 and DM1200 when processing other HLW simulants. Melter DFs were determined for most elements in the feed for all eight test segments performed. The most volatile species were sulfur and fluorine, which is typical. Particulate emissions from the melter increased with increasing glass pool temperature and bubbling rate but no obvious change was observed with feed composition. Gaseous emissions of nitrogen oxides and byproducts of incomplete combustion, such as carbon monoxide and ammonia, were very low due to the lack of nitrates and organic carbon in the feed.

The new glass formulations meet all processing and product quality requirements for each of the developed compositions, with waste loadings ranging from 43.6 to 45 wt% on a waste oxide basis. Glass samples from the crucible and melter tests were subjected to leach testing using the PCT and TCLP methods in order to evaluate product quality. Despite the high waste loadings, the glass products significantly out-performed the DWPF-EA benchmark glass on the PCT leaching procedure by factors of at least 140 and exhibited TCLP leachate concentrations that were well below the WTP delisting limits. Overall, the results from the melter tests and the associated processing and product quality data support the viability of the proposed HLW enhancement approaches. In addition, the results from the DM100 and DM1200 melter tests provided validation of the VGF and DFC melt rate screening tests that were developed and successfully integrated into the glass formulation methodology.

The characteristics of the HWI-Al-19 formulation developed in this work and tested on both the DM100 melter and the DM1200 pilot melter include:

- 45 wt% waste loading
- Limited by crystallization on CCC and impact on PCT; also spinel formation
- $Al_2O_3$  content = 23.97 wt% vs. WTP contract minimum of 11 wt%, giving a 54% reduction in the volume of glass produced from this waste
- Melt rate significantly exceeds the WTP target of 800 kg/m<sup>2</sup>/d (corresponding to 3 MT/d per WTP HLW melter) under nominal conditions
- Melt rate of 1500 kg/m<sup>2</sup>/d demonstrated with optimized bubbling, representing an 87% increase over the WTP target rate.

In view of the importance of aluminum limited waste streams at Hanford (and also Savannah River), the ability to achieve high waste loadings without adversely impacting melt rates has the potential for enormous cost savings from reductions in canister count and the potential for schedule acceleration. Consequently, the potential return on the investment made in the development of these enhancements is extremely favorable.

### 8.1 **Recommendations for Future Work**

The results of the testing presented herein clearly demonstrate the viability of the enhancement and optimization strategies that have been developed. Furthermore, the results have the potential for enormous savings in cost and schedule. As a result, it is recommended that testing and evaluation of these strategies be continued in order to provide a solid basis for their broad implementation in order to maximize the cost and schedule benefits while minimizing technical risk. Some of the elements of such a program are summarized below. While the discussion is centered around Hanford HLW streams, many of these enhancement strategies would also be applicable to other HLW streams, such as those being treated at the DWPF, with the potential for similarly large benefits.

- Other WTP HLW Waste Types: The present testing was based on a single high aluminum HLW composition from the Hanford tanks. While these results are also relevant to waste from several tanks, the diversity of the Hanford tank wastes means that there are several such waste-loading-limiting constituents (and combinations thereof), each of which needs to be addressed in order to maximize the benefit from these enhancements. As a starting point, a similar test program to the one performed in the present work should be performed for each of the principal waste-loading-limiting constituents. Particular attention should be paid to waste streams containing sulfur due to the potential of forming secondary phases on the melt pool surface. In addition, production rates and feed rheology are affected by the form of the major constituents in the waste stream. Melter tests conducted on likely forms of major waste constituents would permit more accurate estimates of potential production rates and processing problems.
- *Throughput*: A key risk area addressed in the present work relates to the strong dependence of glass production rates on waste composition. The extent of this variation across the full spectrum of HLW waste types needs to be quantified in order to accurately project waste treatment rates. The vast majority of the previous testing for the WTP has been performed on iron-limited wastes and those results have formed the basis for vitrification system capacity projections. The results of the present work suggest that this basis may not be appropriate for other waste types. Testing also needs to be performed to develop approaches to mitigate the potential short-fall in glass production rates.
- *Formulating Glasses with higher crystal contents:* Previous tests with HLW iron-limited wastes showed that allowing a higher crystal content product can allow significantly increased waste loadings. Evaluation of this enhanced "operational liquidus temperature" approach for other waste streams would result in further waste loading increases.
- *Scale-Up Testing*: As in the present work, testing should be extended to larger-scale melter systems in order to address potential risks associated with scale-up, particularly with respect to processing rates. Testing should be conducted at the DM1200 WTP HLW Pilot Melter scale (1.2 m<sup>2</sup>). Optimization of bubbling rate is a critical variable and therefore testing with bubblers in the prototypical orientation at larger scale is required to confirm these findings.

- *Integrated System Testing*: Testing on the DM1200 WTP HLW Pilot Melter system provides data from a one-third scale system with a prototypical feed delivery system and off-gas treatment train. Such testing is necessary to evaluate potential interactive effects on system operation arising from implementation of the enhancement strategies and to provide data on the performance of each unit operation, input for flow-sheet models and regulatory requirements, and information of recycle streams.
- *Longer-Duration Testing*: After validated at larger scale, the duration of testing should be extended in order to address and quantify any chronic issues, such as the slow accumulation of crystals in the melter cavity, any degradation in the ability to discharge glass, and effects on off-gas line plugging.

Finally, it is likely that the maximum production rate for each set of test conditions would have been significantly impacted for most of the tests if the cold cap conditions were not monitored visually. Consequently, it is recommended that the ability to maintain production rates without use of visual information be evaluated, if that is the planned WTP operating mode.

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Melt Rate Enhancement for Hig	gh Aluminum HLW Glass Formulations
	Final Report, VSL-08R1360-1, Rev. 0

Waste Component	Bi Limited	Cr Limited	Al Limited	Al and Na Limited
Al <sub>2</sub> O <sub>3</sub>	22.45%	25.53%	49.21%	43.30%
$B_2O_3$	0.58%	0.53%	0.39%	0.74%
CaO	1.61%	2.47%	2.21%	1.47%
Fe <sub>2</sub> O <sub>3</sub>	13.40%	13.13%	12.11%	5.71%
Li <sub>2</sub> O	0.31%	0.36%	0.35%	0.15%
MgO	0.82%	0.16%	0.24%	0.44%
Na <sub>2</sub> O	12.97%	20.09%	7.35%	25.79%
SiO <sub>2</sub>	12.04%	10.56%	10.05%	6.22%
TiO <sub>2</sub>	0.30%	0.01%	0.02%	0.35%
ZnO	0.31%	0.25%	0.17%	0.36%
ZrO <sub>2</sub>	0.40%	0.11%	0.81%	0.25%
SO <sub>3</sub>	0.91%	1.52%	0.41%	0.44%
Bi <sub>2</sub> O <sub>3</sub>	12.91%	7.29%	2.35%	2.35%
ThO <sub>2</sub>	0.25%	0.04%	0.37%	0.04%
Cr <sub>2</sub> O <sub>3</sub>	1.00%	3.07%	1.07%	1.44%
K <sub>2</sub> O	0.89%	0.37%	0.29%	1.34%
$U_3O_8$	3.48%	7.59%	7.25%	4.58%
BaO	0.02%	0.03%	0.11%	0.06%
CdO	0.00%	0.01%	0.05%	0.02%
NiO	3.71%	1.06%	0.82%	0.20%
PbO	0.48%	0.48%	0.84%	0.18%
$P_2O_5$	9.60%	3.34%	2.16%	4.10%
F-	1.58%	2.00%	1.37%	0.46%
Total	100.00%	100.00%	100.00%	100.00%

### Table 2.1. Oxide Compositions of Limiting Waste Streams (wt%).

Table 2.2. Compositions of the Al-Limited Waste (Oxide Basis) and the HLW Waste Simulant to
Produce 100 kg of Waste Oxides (20 wt% suspended solids) Using Al <sub>2</sub> O <sub>3</sub> as the Aluminum Source.

Al-Limited Waste Composition		Al-Limited HLW	Al-Limited HLW Waste Simulant			
Waste Oxide	Wt%	Starting Materials	Target Weight (kg)			
Al <sub>2</sub> O <sub>3</sub>	49.21%	Al <sub>2</sub> O <sub>3</sub>	49.707			
B <sub>2</sub> O <sub>3</sub>	0.39%	H <sub>3</sub> BO <sub>3</sub>	0.700			
CaO	2.21%	CaO	2.255			
Fe <sub>2</sub> O <sub>3</sub>	12.11%	Fe(OH) <sub>3</sub> (13%) Slurry)	99.643			
Li <sub>2</sub> O	0.35%	Li <sub>2</sub> CO <sub>3</sub>	0.888			
MgO	0.24%	MgO	0.253			
Na <sub>2</sub> O	7.35%	NaOH	4.235			
SiO <sub>2</sub>	10.05%	SiO <sub>2</sub>	10.152			
TiO <sub>2</sub>	0.02%	TiO <sub>2</sub>	0.020			
ZnO	0.17%	ZnO	0.172			
ZrO <sub>2</sub>	0.81%	Zr(OH) <sub>4</sub> ·xH <sub>2</sub> O	2.093			
SO <sub>3</sub>	0.41%	Na <sub>2</sub> SO <sub>4</sub>	0.735			
Bi <sub>2</sub> O <sub>3</sub>	2.35%	Bi <sub>2</sub> O <sub>3</sub>	2.374			
ThO <sub>2</sub>	0.37%	Th Surrogate	0			
Cr <sub>2</sub> O <sub>3</sub>	1.07%	Cr <sub>2</sub> O <sub>3</sub> ·1.5H <sub>2</sub> O	1.273			
K <sub>2</sub> O	0.29%	KNO3	0.632			
$U_3O_8$	7.25%	U Surrogate	0			
BaO	0.11%	BaCO <sub>3</sub>	0.143			
CdO	0.05%	CdO	0.051			
NiO	0.82%	Ni(OH) <sub>2</sub>	1.055			
РЬО	0.84%	РЬО	0.848			
P <sub>2</sub> O <sub>5</sub>	2.16%	FePO <sub>4</sub> ·xH <sub>2</sub> O	5.738			
F	1.37%	NaF	3.044			
Carbonate	$1.20^{\#}$	Na <sub>2</sub> CO <sub>3</sub>	0.806			
Nitrite	0.50	NaNO <sub>2</sub>	0.769			
Nitrate		NaNO <sub>3</sub>	2.230			
Organic Carbon	0.05	$H_2C_2O_4$ ·2 $H_2O$	0.264			
_	_	Water	279.400			
— TOTAL	100.0%		469.478			

\* Target weights adjusted for assay information of starting materials # Unit for volatile components is g/100 g of waste oxide

Al-Limited Was	te Composition	Al-Limited HLW	Al-Limited HLW Waste Simulant			
Waste Oxide	Wt%	Starting Materials	Target Weight (kg) <sup>*</sup>			
$Al_2O_3$	49.21%	Al(OH) <sub>3</sub>	76.052			
$B_2O_3$	0.39%	H <sub>3</sub> BO <sub>3</sub>	0.700			
CaO	2.21%	CaO	2.255			
Fe <sub>2</sub> O <sub>3</sub>	12.11%	Fe(OH) <sub>3</sub> (13% Slurry)	99.643			
Li <sub>2</sub> O	0.35%	Li <sub>2</sub> CO <sub>3</sub>	0.888			
MgO	0.24%	MgO	0.253			
Na <sub>2</sub> O	7.35%	NaOH	4.235			
SiO <sub>2</sub>	10.05%	SiO <sub>2</sub>	10.152			
TiO <sub>2</sub>	0.02%	TiO <sub>2</sub>	0.020			
ZnO	0.17%	ZnO	0.172			
ZrO <sub>2</sub>	0.81%	Zr(OH) <sub>4</sub> ·xH <sub>2</sub> O	2.093			
$SO_3$	0.41%	Na <sub>2</sub> SO <sub>4</sub>	0.735			
Bi <sub>2</sub> O <sub>3</sub>	2.35%	Bi <sub>2</sub> O <sub>3</sub>	2.374			
ThO <sub>2</sub>	0.37%	Th Surrogate	0			
Cr <sub>2</sub> O <sub>3</sub>	1.07%	Cr <sub>2</sub> O <sub>3</sub> ·1.5H <sub>2</sub> O	1.273			
K <sub>2</sub> O	0.29%	KNO3	0.632			
$U_3O_8$	7.25%	U Surrogate	0			
BaO	0.11%	BaCO <sub>3</sub>	0.143			
CdO	0.05%	CdO	0.051			
NiO	0.82%	Ni(OH) <sub>2</sub>	1.055			
PbO	0.84%	PbO	0.848			
$P_2O_5$	2.16%	FePO <sub>4</sub> ·xH <sub>2</sub> O	5.738			
F	1.37%	NaF	3.044			
Carbonate	$1.20^{\#}$	Na <sub>2</sub> CO <sub>3</sub>	0.806			
Nitrite	0.50	NaNO <sub>2</sub>	0.769			
Nitrate	2.00	NaNO <sub>3</sub>	2.230			
Organic Carbon	0.05	$H_2C_2O_4 \cdot 2H_2O$	0.264			
		Water	$279.400^4$			
		—				
TOTAL	100.0%	TOTAL	495.825 <sup>4</sup>			

## Table 2.3. Compositions of the Al-Limited Waste (Oxide Basis) and the HLW Waste Simulant to Produce 100 kg of Waste Oxides (20 wt% suspended solids) Using Al(OH)<sub>3</sub> as the Aluminum Source.

\* Target weights adjusted for assay information of starting materials

<sup>#</sup>Unit for volatile components is g/100 g of waste oxide

Table 2.4. Composition and Properties of Aluminum Limited Waste and HLW-E-Al-27 Glass
Formulation at 45% Waste Loading Used in Previous Melter Tests (wt%).

-	Al-Limited Waste*	Waste in Glass	Glass Forming Additives	Target Glass HLW-E-Al-27
Al <sub>2</sub> O <sub>3</sub>	53.27	23.97	-	23.97
B <sub>2</sub> O <sub>3</sub>	0.42	0.19	15.00	15.19
BaO	0.12	0.05	-	0.05
Bi <sub>2</sub> O <sub>3</sub>	2.54	1.14	-	1.14
CaO	2.39	1.08	5.00	6.08
CdO	0.05	0.02	-	0.02
Cr <sub>2</sub> O <sub>3</sub>	1.16	0.52		0.52
F	1.48	0.67	-	0.67
Fe <sub>2</sub> O <sub>3</sub>	13.11	5.90	-	5.90
K <sub>2</sub> O	0.31	0.14	-	0.14
Li <sub>2</sub> O	0.38	0.17	3.40	3.57
MgO	0.26	0.12	-	0.12
Na <sub>2</sub> O	7.96	3.58	6.00	9.58
NiO	0.89	0.40	-	0.40
$P_2O_5$	2.34	1.05	-	1.05
PbO	0.91	0.41	-	0.41
SiO <sub>2</sub>	10.88	4.90	25.60	30.50
TiO <sub>2</sub>	0.02	0.01	-	0.01
SO <sub>3</sub>	0.44	0.20	-	0.20
ZnO	0.18	0.08	-	0.08
ZrO <sub>2</sub>	0.88	0.39	-	0.39
Sum	100.00	45.00	55.00	100.00

\* Renormalized from Ref. [5] after removal of radioactive components

Viscos	Viscosity @1150°C, P				
Conductiv	Conductivity @1150°C, S/cm				
Crystal C	ontent, As Melted	Trace			
Crystal Cor	Crystal Content, 72 hr at 950°C				
Crystal	Content, CCC	~1.9 vol%			
	Pass				
	-	DWPF-EA	HLW-E-Al-27		
DCT o/I	В	16.7	0.27		
PCT, g/L	Li	9.6	0.44		
	Na	13.3	0.30		

## Table 2.5. Compositions of Melter Feed to Produce 100 kg of Target Glass HLW-E-Al-27 (Glass Yield = 500 g/L Feed) from the Al-Limited Waste Simulant Using Al(OH)<sub>3</sub> as the Aluminum Source.

Al-Limited W	Al-Limited Waste Simulant		ng Additives
Starting Materials	Target Weight (kg) <sup>*</sup>	Starting Materials	Target Weight (kg)*
Al(OH) <sub>3</sub>	37.047		_
H <sub>3</sub> BO <sub>3</sub>	0.341	Na <sub>2</sub> B <sub>4</sub> O <sub>7</sub> ·10H <sub>2</sub> O	37.292
BaCO <sub>3</sub>	0.070	H <sub>3</sub> BO <sub>3</sub>	2.729
Bi <sub>2</sub> O <sub>3</sub>	1.156		
CaO	1.099	CaSiO <sub>3</sub> (Wollastonite)	10.886
CdO	0.025	— —	
Cr <sub>2</sub> O <sub>3</sub>	0.532		
NaF	1.483		
Fe(OH) <sub>3</sub> (13% Slurry)	48.539	—	
KNO3	0.308		
Li <sub>2</sub> CO <sub>3</sub>	0.432	Li <sub>2</sub> CO <sub>3</sub>	8.625
MgO	0.121	—	
NaOH	2.190		
Ni(OH) <sub>2</sub>	0.514	—	
FePO <sub>4</sub> ·xH <sub>2</sub> O	2.795		
PbO	0.413	—	
$Na_2SO_4$	0.358	—	
SiO <sub>2</sub>	4.945	SiO <sub>2</sub>	20.251
TiO <sub>2</sub>	0.010		
ZnO	0.084		
Zr(OH) <sub>4</sub> ·xH <sub>2</sub> O	1.020	—	
H <sub>2</sub> O	105.479	—	
Na <sub>2</sub> CO <sub>3</sub>	0.314	_	
NaNO <sub>2</sub>	0.346	—	
NaNO <sub>3</sub>	0.984	—	
$H_2C_2O_4$ ·2 $H_2O$	0.119	—	
		—	
Simulant Total	210.724	Additives Total	79.783
		FEED TOTAL	290.507

\*Target weights adjusted for assay information of starting materials

# Table 3.1. Summary of Glass Formulation and the Associated Feed Tests by VGF and Waste Glass Melters (DM100 or DM1200).

Glass Name	Feed Name	Aluminum Source in feed	Boron Source in feed	VGF Test	DM100 Test	DM1200 Test
HLW-E-Al-27	BLL-F-135A [2]	Al <sub>2</sub> O <sub>3</sub>	Borax	Yes	Yes	No
HLW-E-Al-27	BLX-F-50	Al(OH) <sub>3</sub>	Borax	Yes	Yes	No
HLW-E-Al-27	F-Al-VAR-1	Al <sub>2</sub> O <sub>3</sub>	Boric acid	Yes	No	No
HLW-E-Al-27	F-Al-VAR-4	Al(OH) <sub>3</sub>	Borax	Yes	No	No
HWL-E-Al-27	F-Al-VAR-2	Al(OH) <sub>3</sub>	Boric acid	Yes	No	No
HLW-E-ANa-22	BLN-F-21A [2]	Al <sub>2</sub> O <sub>3</sub>	Boric Acid	Yes	Yes	No
HLW-E-Bi-6	BLK-F-84A[2]	Al <sub>2</sub> O <sub>3</sub>	Borax	Yes	Yes	No
HLW-E-CrM	BLL-F-65A[2]	$Al_2O_3$	Boric acid	Yes	Yes	No
Matrix 1-B1	BLS-F-7A [24]	$Al_2O_3$	Borax	Yes	Yes	No
Matrix 2-9	BLV-F-132B [25]	Al <sub>2</sub> O <sub>3</sub>	Borax	Yes	Yes	No
HWI-Al-1	F-HWI-Al-1B	Al <sub>2</sub> O <sub>3</sub>	Boric Acid	Yes	No	No
HWI-Al-2	-	-	-	-	-	-
HWI-Al-3	-	-	-	-	-	-
HWI-Al-4	-	-	-	-	-	-
HWI-Al-5	F-HWI-Al-5	Al <sub>2</sub> O <sub>3</sub>	Boric Acid	Yes	No	No
HWI-Al-6	-	Al <sub>2</sub> O <sub>3</sub>	Boric Acid	-	-	-
HWI-Al-7	F-HWI-Al-7	Al <sub>2</sub> O <sub>3</sub>	Boric Acid	Yes	No	No
HWI-Al-8	-	-	-	-	-	-
HWI-Al-9	F-HWI-Al-9	Al <sub>2</sub> O <sub>3</sub>	Boric Acid	Yes	No	No
HWI-Al-10	-	-	-	-	-	-
HWI-Al-11	-	-	-	-	-	-
HWI-Al-12	-	-	-	-	-	-
HWI-Al-13	F-HWI-Al-13	Al <sub>2</sub> O <sub>3</sub>	Boric Acid	Yes	No	No
HWI-Al-14	-	-	-	-	-	-
HWI-Al-15	-	-	-	-	-	-
HWI-Al-16	F-HWI-Al-16	Al <sub>2</sub> O <sub>3</sub>	Boric Acid	Yes	Yes	No
HWI-Al-16	F-HWI-Al-16B	Al(OH) <sub>3</sub>	Boric Acid	Yes	Yes	No
HWI-Al-17	F-HWI-Al-17	Al <sub>2</sub> O <sub>3</sub>	Boric Acid	Yes	No	No
HWI-Al-18	F-HWI-Al-18	Al <sub>2</sub> O <sub>3</sub>	Boric Acid	Yes	No	No
HWI-Al-19	F-HWI-Al-19F	Al <sub>2</sub> O <sub>3</sub>	Boric Acid	Yes	No	No
HWI-Al-19	F-HWI-Al-19B	Al(OH) <sub>3</sub>	Boric Acid	Yes	Yes	Yes
HWI-A1-20	F-HWI-Al-9KSM	Al <sub>2</sub> O <sub>3</sub>	Boric Acid	Yes	No	No

Glass Name	Al-Limited Waste [2]	HLW-E-A1-27 [2]	HWI-Al-1		HWI-A1-2		HWI-	Al-3
Waste loading %	100	45	45		45		42.5	
-	Target	Target	Target	XRF	Target	XRF	Target	XRF
$Al_2O_3$	53.27	23.97	23.97	22.16	23.97	22.14	22.64	20.88
$B_2O_3$	0.42	15.19	18.19	18.19	17.69	17.69	18.18	18.18
BaO	0.12	0.05	0.05	0.06	0.05	0.05	0.05	0.05
Bi <sub>2</sub> O <sub>3</sub>	2.54	1.14	1.14	1.29	1.14	1.21	1.08	1.17
CaO	2.39	6.08	1.08	1.21	1.08	1.13	1.02	1.09
CdO	0.05	0.02	0.02	_	0.02	_	0.02	_
Cr <sub>2</sub> O <sub>3</sub>	1.16	0.52	0.52	0.68	0.52	0.64	0.49	0.62
F	1.48	0.67	0.67	NA	0.67	NA	0.63	NA
Fe <sub>2</sub> O <sub>3</sub>	13.11	5.90	5.90	6.26	5.90	5.85	5.57	5.65
K <sub>2</sub> O	0.31	0.14	0.14	0.14	0.14	0.14	0.13	0.12
Li <sub>2</sub> O	0.38	3.57	4.57	4.57	2.17	2.17	3.16	3.16
MgO	0.26	0.12	0.12	0.09	0.12	0.07	0.11	0.07
Na <sub>2</sub> O	7.96	9.58	9.58	9.80	12.58	13.86	13.38	14.19
NiO	0.89	0.40	0.40	0.42	0.40	0.37	0.38	0.39
$P_2O_5$	2.34	1.05	1.05	1.24	2.05	2.34	1.99	2.30
PbO	0.91	0.41	0.41	0.41	0.41	0.37	0.39	0.38
SiO <sub>2</sub>	10.88	30.50	31.50	32.66	30.40	31.22	30.12	30.99
TiO <sub>2</sub>	0.02	0.01	0.01	0.03	0.01	0.02	0.01	0.02
SO <sub>3</sub>	0.44	0.20	0.20	0.23	0.20	0.22	0.19	0.23
ZnO	0.18	0.08	0.08	0.12	0.08	0.10	0.08	0.10
ZrO <sub>2</sub>	0.88	0.39	0.39	0.38	0.39	0.35	0.37	0.36
Sum	100.00	100.00	100.00	99.92	100.00	99.94	100.00	99.94

Table 5.2. Target and XKF Analysis of HWT Glasses Tested (wt%) (continued).											
Glass Name	HWI-	Al-4	HWI-	Al-5	HWI-	Al-6	Н	WI-Al-7	7	HWI-	Al-8
Waste loading %	45		45	5	45		45			45	
-	Target	XRF	Target	XRF	Target	XRF	Target	XRF	DCP	Target	XRF
$Al_2O_3$	23.97	23.56	23.97	23.33	23.97	23.51	23.97	23.64	22.07	23.97	23.51
$B_2O_3$	17.69	17.69	15.19	15.19	15.19	15.19	15.19	15.19	16.51	15.19	15.19
BaO	0.05	0.05	0.05	0.06	0.05	0.06	0.05	0.05	0.06	0.05	0.05
Bi <sub>2</sub> O <sub>3</sub>	1.14	1.22	1.14	1.22	1.14	1.22	1.14	1.18	NA	1.14	1.15
CaO	1.08	1.11	1.08	1.18	1.08	1.12	3.08	3.24	3.05	1.08	1.11
CdO	0.02	-	0.02	-	0.02	-	0.02	-	NA	0.02	-
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.63	0.52	0.63	0.52	0.64	0.52	0.63	0.46	0.52	0.62
F	0.67	NA	0.67	NA	0.67	NA	0.67	NA	NA	0.67	NA
Fe <sub>2</sub> O <sub>3</sub>	5.90	5.85	5.90	5.86	5.90	5.84	5.90	5.75	5.82	5.90	5.71
K <sub>2</sub> O	4.14	3.82	0.14	0.14	6.14	5.94	0.14	0.13	0.17	8.14	8.03
Li <sub>2</sub> O	2.17	2.17	3.57	3.57	2.57	2.57	3.57	3.57	4.08	2.17	2.17
MgO	0.12	-	5.12	4.63	0.12	0.01	3.12	2.88	3.07	0.12	0.09
Na <sub>2</sub> O	9.58	9.95	9.58	10.42	9.58	9.91	9.58	9.78	8.91	9.08	9.44
NiO	0.40	0.37	0.40	0.39	0.40	0.38	0.40	0.37	0.38	0.40	0.36
P <sub>2</sub> O <sub>5</sub>	1.05	1.23	1.05	1.19	1.05	1.21	1.05	1.19	1.10	1.05	1.24
PbO	0.41	0.37	0.41	0.39	0.41	0.38	0.41	0.38	NA	0.41	0.37
SiO <sub>2</sub>	30.40	31.21	30.50	30.97	30.50	31.17	30.50	31.27	33.72	29.40	30.20
TiO <sub>2</sub>	0.01	0.02	0.01	0.02	0.01	0.02	0.01	0.02	0.03	0.01	0.02
$SO_3$	0.20	0.21	0.20	0.26	0.20	0.23	0.20	0.24	NA	0.20	0.23
ZnO	0.08	0.10	0.08	0.10	0.08	0.10	0.08	0.10	NA	0.08	0.10
ZrO <sub>2</sub>	0.39	0.36	0.39	0.38	0.39	0.37	0.39	0.36	0.49	0.39	0.35
Sum	100.00	99.92	100.00	99.94	100.00	99.87	100.00	99.97	99.92	100.00	99.94
NA – Not analyze	NA – Not analyzed										

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Glass Name	HWI-Al-9			Н	HWI-Al-10			HWI-Al-11		HWI-Al-12		HWI-Al-13	
Waste Loading %	40			40			45		45		45		
-	Target	XRF	DCP	Target	XRF	DCP	Target	XRF	Target	XRF	Target	XRF	
$Al_2O_3$	21.31	20.76	19.46	21.31	20.62	19.42	23.97	23.36	23.97	23.46	23.97	23.41	
$B_2O_3$	18.17	18.17	19.42	18.67	18.67	19.98	18.19	18.19	18.19	18.19	15.19	15.19	
BaO	0.05	0.06	0.05	0.05	0.06	0.05	0.05	0.05	0.05	0.05	0.05	0.06	
Bi <sub>2</sub> O <sub>3</sub>	1.02	1.07	NA	1.02	1.12	NA	1.14	1.18	1.14	1.17	1.14	1.21	
CaO	0.96	1.00	1.21	0.96	1.04	1.20	3.08	3.20	2.08	2.14	6.08	6.24	
CdO	0.02	-	NA	0.02	١	NA	0.02	-	0.02	_	0.02	-	
Cr <sub>2</sub> O <sub>3</sub>	0.46	0.59	0.50	0.46	0.60	0.50	0.52	0.63	0.52	0.62	0.52	0.63	
F	0.59	NA	NA	0.59	NA	NA	0.67	NA	0.67	NA	0.67	NA	
Fe <sub>2</sub> O <sub>3</sub>	5.24	5.20	5.27	5.24	5.39	5.32	5.90	5.82	5.90	5.71	5.90	5.77	
K <sub>2</sub> O	0.13	0.12	0.17	0.13	0.12	0.18	0.14	0.13	0.14	0.14	4.14	4.07	
Li <sub>2</sub> O	3.15	3.15	3.55	3.65	3.65	4.05	3.57	3.57	3.57	3.57	3.57	3.57	
MgO	0.10	0.07	0.13	0.10	0.08	0.12	0.12	0.07	0.12	0.11	0.12	0.10	
Na <sub>2</sub> O	13.18	13.89	12.07	13.18	13.31	12.12	9.58	9.88	9.58	10.18	6.08	6.65	
NiO	0.36	0.35	0.38	0.36	0.37	0.38	0.40	0.37	0.40	0.36	0.40	0.38	
$P_2O_5$	0.94	1.06	1.03	0.94	1.06	1.02	1.05	1.24	1.05	1.22	1.05	1.19	
PbO	0.36	0.34	NA	0.36	0.36	NA	0.41	0.37	0.41	0.37	0.41	0.38	
SiO <sub>2</sub>	33.35	33.48	36.31	32.35	32.82	35.39	30.50	31.19	31.50	32.01	30.00	30.39	
TiO <sub>2</sub>	0.01	0.02	0.02	0.01	0.02	0.03	0.01	0.02	0.01	0.02	0.01	0.02	
$SO_3$	0.18	0.22	NA	0.18	0.23	NA	0.20	0.24	0.20	0.23	0.20	0.25	
ZnO	0.07	0.09	NA	0.07	0.10	NA	0.08	0.10	0.08	0.08	0.08	0.09	
ZrO <sub>2</sub>	0.35	0.34	0.42	0.35	0.35	0.44	0.39	0.35	0.39	0.34	0.39	0.37	
Sum	100.00	99.96	99.99	100.00	99.96	100.20	100.00	99.96	100.00	99.97	100.00	99.97	

NA – Not analyzed - Empty data field

Glass Name	HWI-AL-14		HWI-A	I-AL-15 HWI-A		Al-16	HWI-Al-17		HWI-Al-18		HWI-Al-18G*	
Waste Loading	45		45		43.6		43.6		42.		.4	
-	Target	XRF	Target	XRF	Target	XRF	Target	XRF	Target	XRF	XRF	
$Al_2O_3$	23.97	23.60	23.97	23.49	23.25	22.71	23.25	22.50	22.57	21.89	21.73	
B <sub>2</sub> O <sub>3</sub>	15.19	15.19	13.19	13.19	17.73	17.73	14.73	14.73	20.13	20.13	20.13	
BaO	0.05	0.07	0.05	0.07	0.05	0.06	0.05	0.05	0.05	0.04	0.06	
Bi <sub>2</sub> O <sub>3</sub>	1.14	1.19	1.14	1.17	1.11	1.16	1.11	1.21	1.08	1.13	0.95	
CaO	5.08	5.21	5.08	5.16	5.89	6.05	5.89	6.06	5.72	5.94	5.76	
CdO	0.02	-	0.02	-	0.02	-	0.02	-	0.02	-	NA	
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.64	0.52	0.64	0.51	0.62	0.51	0.64	0.49	0.61	0.54	
F	0.67	NA	0.67	NA	0.65	NA	0.65	NA	0.63	NA	NA	
Fe <sub>2</sub> O <sub>3</sub>	5.90	5.72	5.90	5.72	5.72	5.59	5.72	5.80	5.56	5.58	5.69	
K <sub>2</sub> O	5.14	4.96	5.14	5.06	0.14	0.13	3.14	3.07	0.13	0.16	0.16	
Li <sub>2</sub> O	3.57	3.57	3.17	3.17	3.46	3.46	3.46	3.46	3.36	3.36	3.36	
MgO	0.12	0.10	0.12	-	0.11	0.10	0.11	0.07	0.11	0.08	0.31	
Na <sub>2</sub> O	6.08	6.14	8.58	8.97	9.29	9.68	9.29	10.05	9.02	9.60	9.28	
NiO	0.40	0.38	0.40	0.37	0.39	0.37	0.39	0.39	0.38	0.37	0.36	
P <sub>2</sub> O <sub>5</sub>	1.05	1.24	1.05	1.20	1.02	1.16	1.02	1.13	0.99	1.11	1.11	
PbO	0.41	0.37	0.41	0.37	0.40	0.36	0.40	0.39	0.39	0.35	0.35	
SiO <sub>2</sub>	30.00	30.86	29.90	30.58	29.58	30.09	29.58	29.67	28.72	28.91	29.33	
TiO <sub>2</sub>	0.01	0.04	0.01	0.02	0.01	0.02	0.01	0.02	0.01	0.02	0.04	
SO <sub>3</sub>	0.20	0.22	0.20	0.25	0.19	0.23	0.19	0.23	0.19	0.23	0.20	
ZnO	0.08	0.10	0.08	0.10	0.08	0.10	0.08	0.10	0.08	0.11	0.08	
ZrO <sub>2</sub>	0.39	0.36	0.39	0.38	0.38	0.35	0.38	0.38	0.37	0.33	0.48	
Sum	100.00	99.96	100.00	99.91	100.00	99.97	100.00	99.96	100.00	99.94	99.91	

Table 3.2. Target and XRF Analysis of HWI Glasses Tested (wt%) (continued).

\* Identical to HWI-Al-18, melted from melter feed and boron additives

NA – Not analyzed

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Glass Name	HWI-2	Al-19	HWI-Al-20				
Waste Loading %	4:	5	45				
-	Target	XRF	Target	XRF	DCP		
$Al_2O_3$	23.97	22.94	23.97	23.30	22.00		
$B_2O_3$	19.19	19.19	18.19	18.19	18.85		
BaO	0.05	0.06	0.05	0.07	0.07		
Bi <sub>2</sub> O <sub>3</sub>	1.14	1.34	1.14	1.20	1.30		
CaO	5.58	6.06	1.08	1.16	1.28		
CdO	0.02	-	0.02	_	NA		
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.69	0.52	0.64	0.50		
F	0.67	NA	0.67	NA	NA		
Fe <sub>2</sub> O <sub>3</sub>	5.90	6.32	5.90	5.82	5.79		
K <sub>2</sub> O	0.14	0.16	5.14	5.26	4.61		
Li <sub>2</sub> O	3.57	3.57	3.17	3.17	3.58		
MgO	0.12	0.09	0.12	-	0.44		
Na <sub>2</sub> O	9.58	9.54	8.58	9.48	8.21		
NiO	0.40	0.42	0.40	0.38	0.39		
$P_2O_5$	1.05	1.22	1.05	1.21	1.09		
PbO	0.41	0.43	0.41	0.39	NA		
SiO <sub>2</sub>	27.00	27.10	28.90	28.94	31.16		
TiO <sub>2</sub>	0.01	0.03	0.01	0.01	0.05		
SO <sub>3</sub>	0.20	0.23	0.20	0.22	NA		
ZnO	0.08	0.12	0.08	0.10	0.08		
ZrO <sub>2</sub>	0.39	0.42	0.39	0.38	0.53		
Sum	100.00	99.92	100.00	99.91	99.93		

### Table 3.2. Target and XRF Analysis of HWI Glasses Tested (wt%) (continued).

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	Formulation Group	1	2	2	4	3
	Sample ID	HWI-Al-1	HWI-Al-2	HWI-Al-3	HWI-Al-4	HWI-Al-5
	Waste loading %	45	45	42.5	45	45
	Feed melt rate rank (1-6)**	2	NA	NA	NA	6
	Glass as melted	Minor PS*	Minor PS	Minor PS	PS	Good
Crystal content	Crystals at 950°C	Spinel	-	-	-	Spinel
	After heat treatment at 950°C (vol%)	1.7	_	-	-	3.4
	After heat treatment at 900°C (vol%)	1.6	-	-	-	
	After heat treatment at 850°C (vol%)	1.4	-	-	-	
C	After heat treatment at 800°C (vol%)	2.3	-	-	-	5.3
	After CCC heat treatment (vol%)	-	_	-	-	5.2
	Viscosity at 1250°C (Poise)	24.4	_	-	-	-
	Viscosity at 1150°C (Poise)	55.1	_	-	-	-
S	Viscosity at 1050°C (Poise)	142.3	_	-	-	-
ertie	Viscosity at 950°C (Poise)	436.9	-	-	-	-
rop	Electric conductivity at 1250°C					
d St	(S/cm)	0.42	-	-	-	-
SSil	Electric conductivity at 1150°C (S/cm)	0.32	_	_		_
Processing properties	Electric conductivity at 1050°C	0.52				
P	(S/cm)	0.23	-	-	-	-
	Electric conductivity at 950°C (S/cm)	0.15	-	-	-	-
	Glass transition temperature (°C)	-	-	-	-	-
	РСТ	-	-	-	-	-
0	B ( g/L)	-	-	-	-	-
ance	Li (g/L)	-	-	-	-	-
L	$\frac{\text{Na}(\text{g/L})}{\sum_{i=1}^{n} (i - i)}$	-	-	-	-	-
erfc	Si (g/L) PCT (for CCC sample)	-	-	-	-	-
Leaching performance	B (g/L)	-	-	-	-	-
chiı	Li (g/L)		_	_	_	_
Lea	Na (g/L)	-	-	-	-	-
	Si (g/L)	-	-	-	-	-
	TCLP (Pass/Fail)	-	-	-	PASS	-

#### Table 3.3. Characterization of HWI Glasses.

\*PS - Secondary phases present \*\* - Refer to Table 3.5 for definition

	Formulation Group	4	3	4	1	1	5
	Sample ID	HWI-Al-6	HWI-Al-7	HWI-Al-8	HWI-Al-9	HWI-Al-10	HWI-Al-11
	Waste loading %	45	45	45	40	40	45
	Feed melt rate rank (1-6)**	NA	5	NA	2	NA	NA
	Glass as melted	Minor PS	Good	PS	Good	Good	PS
nt	Crystals at 950°C	-	Spinel	-	Spinel	Spinel	-
Crystal content	After heat treatment at 950°C (vol%)	-	2.7	-	0.7	0.4	-
al cc	After heat treatment at 900°C (vol%)	-	2.8	-	0.8	0.8	-
ystá	After heat treatment at 850°C (vol%)	-	3.2	-	0.9	0.9	-
C	After heat treatment at 800°C (vol%)	-	3.9	-	1.1	0.8	-
	After CCC heat treatment (vol%)	-	3.7	-	0.7	1.3	-
	Viscosity at 1250°C (Poise)	-	22.9	-	29.7	-	-
S	Viscosity at 1150°C (Poise)	-	60.6	-	63.2	-	-
ertie	Viscosity at 1050°C (Poise)	-	192.5	-	163.0	-	-
rope	Viscosity at 950°C (Poise)	-	777.5	-	555.9	-	-
ng p	Electric conductivity at 1250°C (S/cm)	-	0.35	-	0.44	-	-
SSir	Electric conductivity at 1150°C (S/cm)	-	0.25	-	0.33	-	-
Processing properties	Electric conductivity at 1050°C (S/cm)	-	0.17	-	0.23	-	-
P	Electric conductivity at 950°C (S/cm)	-	0.11	-	0.14	-	-
	Glass transition temperature (°C) (S/cm)	-	-	-	-	-	-
	РСТ	-	-	-	-	-	-
	B (g/L)	-	0.267	-	1.435	-	-
nce	Li (g/L)	-	0.467	-	1.150	-	-
rma	Na (g/L)	-	0.305	-	0.777	-	-
erfo	Si (g/L)	-	0.201	-	0.311	-	-
g þ(	PCT (for CCC sample) $P(z/L)$	-	-	-	3.501	-	-
Leaching performance	B (g/L) Li (g/L)	-	-	-	2.552	-	-
eac	Na (g/L)	-	-	-	1.383	-	-
	Si (g/L)				0.216		
	TCLP (Pass/Fail)	-	PASS	-	-	-	_

#### Table 3.3. Characterization of HWI Glasses (continued).

\*PS - Secondary phases present \*\* - Refer to Table 3.5 for definition

	Formulation Group	5	4	4	4	5
	Sample ID	HWI-Al-12	HWI-Al-13	HWI-AL-14	HWI-AL-15	HWI-Al-16
	Waste loading %	45	45	45	45	43.6
	Feed melt rate rank (1-6)**	NA	4	NA	NA	3
	Glass as melted	PS	Minor PS	Minor PS	PS	Good
ent	Crystals at 950°C	-	Spinel + minor Apatitite	-	-	Spinel+Apatite
Crystal content	After heat treatment at 950°C (vol%)	-	2.5	-	-	1.6
tal (	After heat treatment at 900°C (vol%)	-	3	-	-	2.2
rys	After heat treatment at 850°C (vol%)	-	3.4	-	-	2.8
0	After heat treatment at 800°C (vol%)	-	4.4	-	-	3.4
	After CCC heat treatment (vol%)	-	2.6	-	-	1.9
	Viscosity at 1250°C (Poise)	-	-	-	-	18.9
s	Viscosity at 1150°C (Poise)	-	-	-	-	43.1
ertie	Viscosity at 1050°C (Poise)	-	-	-	-	119.1
rope	Viscosity at 950°C (Poise)	-	_	-	-	432.2
Processing properties	Electric conductivity at 1250°C (S/cm)	-	-	-	-	0.35
ssir	Electric conductivity at 1150°C (S/cm)	_	-	-	-	0.24
roce	Electric conductivity at 1050°C (S/cm)	-	-	-	-	0.16
P	Electric conductivity at 950°C (S/cm)	-	-	-	-	0.09
	Glass transition temperature (°C)	-	-	-	-	475
	РСТ	-	_	-	_	_
	B ( g/L)	-	_	-	-	0.314
nce	Li (g/L)	-	-	-	-	0.394
rma	Na (g/L)	-	-	-	-	0.285
erfo	Si (g/L)	-	-	-	-	0.134
g pe	PCT (for CCC sample)	-	-	-	-	-
Leaching performance	B (g/L) Li (g/L)	-	0.200	-	-	0.263 0.329
leac	Na (g/L)	-	0.177	-	-	0.329
	Si (g/L)	_	0.120	-	_	0.132
	TCLP (Pass/Fail)	-	-	-	-	PASS

#### Table 3.3. Characterization of HWI Glasses (continued).

\*PS - Secondary phases present

\*\* - Refer to Table 3.5 for definition

	Formulation Group	4	5	5	5	4
	Sample ID	HWI-Al-17	HWI-Al-18	HWI-Al-18G	HWI-Al-19	HWI-Al-20
	Waste loading %	43.6	42.4	42.4	45	45
	Feed melt rate rank (1-6)**	4	NA	2	3	2
	Glass as melted	Minor PS	Good	Good	Good	PS
int	Crystals at 950°C	Spinel + minor Apatite	-	Spinel	Spinel	Spinel
onte	After heat treatment at 950°C (vol%)	1.1	-	1.1	1.3	1.6
al co	After heat treatment at 900°C (vol%)	1.9	-	2.4	2.1	1.9
Crystal content	After heat treatment at 850°C (vol%)	2.8	-	2.8	3	2.1
Ū	After heat treatment at 800°C (vol%)	3.3	-	3.6	3.7	2.4
	After CCC heat treatment (vol%)	1.3	-	1.8	1.9	3.3
	Viscosity at 1250°C (Poise)	-	-	19.4	14.9	34.4
S	Viscosity at 1150°C (Poise)	-	_	43.8	33.3	78.3
ertie	Viscosity at 1050°C (Poise)	-	-	123.0	90.8	215.0
rope	Viscosity at 950°C (Poise)	-	-	477.6	327.1	765.3
Processing properties	Electric conductivity at 1250°C (S/cm)	-	-	0.29	0.38	0.33
SSir	Electric conductivity at 1150°C(S/cm)	-	-	0.21	0.27	0.23
roce	Electric conductivity at 1050°C(S/cm)	-	-	0.15	0.17	0.15
P	Electric conductivity at 950°C(S/cm)	-	_	0.09	0.09	0.09
	Glass transition temperature (°C)	-	_	-	461	-
	РСТ	-	-	-	-	-
	B ( g/L)	-	-	0.377	0.654	2.109
nce	Li (g/L)	-	-	0.400	0.794	1.584
rma	Na (g/L)	-	-	0.322	0.624	1.306
irfo	Si (g/L)	-	-	0.163	0.223	0.254
g pe	PCT (for CCC sample)	-	-	-	-	-
Leaching performance	B(g/L)	0.856	-	0.342	0.574	-
eac	Li (g/L)	0.534	-	0.402	0.656	-
Ĺ		0.382	-	0.300	0.536	-
	Si (g/L)	0.064	-	0.179	0.232	
	TCLP (Pass/Fail)		-	PASS	PASS	PASS

#### Table 3.3. Characterization of HWI Glasses (continued).

\*PS - Secondary phases present \*\* - Refer to Table 3.5 for definition

		HLW-E-Al	-27		HLW-E-AN	a-22		HLW-E-Bi-6			HLW-E-CrM		
	Target	VGF (45 min)	Fused (1150°C)	Target	VGF (15 min)	Fused (1150°C)	Target	VGF (45 min)	Fused (1150°C)	Target	VGF (5 min)	Fused (1150°C)	
$Al_2O_3$	23.97	22.20	22.40	21.34	21.04	20.56	11.66	11.04	10.91	8.98	8.76	8.71	
$B_2O_3^*$	15.19	15.19	15.19	18.37	18.37	18.37	11.30	11.30	11.30	16.17	16.17	16.17	
BaO	0.05	0.36	0.35	0.03	0.04	0.05	0.01	0.02	0.02	0.01	0.02	0.02	
Bi <sub>2</sub> O <sub>3</sub>	1.14	1.16	1.24	1.16	1.22	1.22	6.71	6.45	6.63	2.56	2.63	2.69	
CaO	6.08	5.83	6.03	0.72	1.04	0.99	0.84	0.88	0.88	0.87	0.80	0.88	
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.53	0.57	0.71	0.83	0.86	0.52	0.59	0.64	1.08	1.13	1.25	
Fe <sub>2</sub> O <sub>3</sub>	5.90	5.87	6.27	2.82	3.04	3.06	6.96	7.19	7.41	4.62	4.59	4.75	
K <sub>2</sub> O	0.14	0.34	0.33	0.66	0.69	0.69	0.46	0.53	0.53	6.05	5.24	5.68	
$Li_2O^*$	3.57	3.57	3.57	3.58	3.58	3.58	0.16	0.18	0.16	3.68	3.68	3.68	
MgO	0.12	0.39	0.42	0.22	0.25	0.23	0.43	0.52	0.43	0.06	0.21	0.22	
Na <sub>2</sub> O	9.58	9.96	9.18	12.71	11.85	12.16	15.74	15.28	15.90	7.07	7.24	7.54	
NiO	0.40	0.35	0.39	0.10	0.12	0.12	1.93	1.72	1.87	0.37	0.47	0.39	
$P_2O_5$	1.05	1.22	1.18	2.02	2.24	2.12	4.99	5.56	5.30	1.18	1.53	1.38	
PbO	0.41	0.35	0.37	0.09	0.10	0.10	0.25	0.29	0.30	0.17	0.15	0.15	
$SO_3$	0.20	0.26	0.20	0.22	0.25	0.22	0.48	0.69	0.52	0.53	0.50	0.44	
SiO <sub>2</sub>	30.50	31.75	31.65	34.56	34.73	35.09	36.26	36.64	36.09	45.76	46.60	45.82	
TiO <sub>2</sub>	0.01	0.04	0.03	0.17	0.18	0.18	0.16	0.19	0.19	0.00	0.03	0.00	
ZnO	0.08	0.08	0.09	0.18	0.18	0.17	0.16	0.19	0.19	0.09	0.10	0.10	
ZrO <sub>2</sub>	0.39	0.46	0.49	0.12	0.18	0.18	0.21	0.39	0.40	0.04	0.07	0.06	

#### Table 3.4a. XRF Analysis (wt%) of Four Melter Feeds Fused at 1150°C and After VGF Experiments.

Glass Name	HWI	-Al-1	HWI	-Al-9
Feed Name	F-HWI-AL-1B- 30M	F-HWI-AL-1B- DHG	F-HWI-AL-9- 30M	F-HWI-AL-9- DHG
Sample Condition	VGF (30 minutes)	Fused at 1150°C	VGF (30 minutes)	Fused at 1150°C
$Al_2O_3$	22.68	23.08	20.79	20.32
$B_2O_3^*$	18.19	18.19	18.17	18.17
BaO	0.07	0.05	0.05	0.06
Bi <sub>2</sub> O <sub>3</sub>	1.23	1.05	1.10	1.12
CaO	1.12	0.99	1.10	1.03
$Cr_2O_3$	0.63	0.60	0.58	0.59
Fe <sub>2</sub> O <sub>3</sub>	5.90	5.56	5.25	5.44
K <sub>2</sub> O	0.17	0.15	0.14	0.14
Li <sub>2</sub> O*	4.57	4.57	3.15	3.15
MgO	0.17	0.34	0.16	0.29
Na <sub>2</sub> O	9.58	10.38	12.80	13.58
NiO	0.38	0.36	0.35	0.40
P <sub>2</sub> O <sub>5</sub>	1.16	1.07	1.03	1.01
PbO	0.38	0.29	0.35	0.37
SO <sub>3</sub>	0.26	0.29	0.24	0.25
SiO <sub>2</sub>	32.87	32.42	34.14	33.44
TiO <sub>2</sub>	0.04	0.05	0.05	0.07
ZnO	0.10	0.08	0.09	0.09
$ZrO_2$	0.43	0.42	0.40	0.45

#### Table 3.4b. XRF Analysis (wt%) of Group 1 Formulation Feeds Fused at 1150°C and After VGF Experiments.

Glass Name	HWI	-Al-5	HWI	-Al-7
Feed Name	F-HWI-AL-5-30M	F-HWI-AL-5-DHG	F-HWI-AL-7-30M	F-HWI-AL-7- DHG
Sample Condition	VGF (30 minutes)	Fused at 1150°C	VGF (30 minutes)	Fused at 1150°C
Al <sub>2</sub> O <sub>3</sub>	21.69	23.13	23.01	23.01
${\rm B_{2}O_{3}}^{*}$	15.19	15.19	15.19	15.19
BaO	0.06	0.05	0.05	0.05
Bi <sub>2</sub> O <sub>3</sub>	1.15	1.25	1.16	1.16
CaO	1.16	1.17	3.16	3.16
$Cr_2O_3$	0.63	0.64	0.64	0.64
$Fe_2O_3$	5.71	6.02	5.57	5.57
K <sub>2</sub> O	0.17	0.16	0.17	0.17
$Li_2O^*$	3.57	3.57	3.57	3.57
MgO	4.68	4.63	2.80	2.80
Na <sub>2</sub> O	10.06	9.59	10.12	10.12
NiO	0.39	0.40	0.38	0.38
$P_2O_5$	1.16	1.13	1.13	1.13
PbO	0.37	0.33	0.87	0.87
$SO_3$	0.34	0.29	0.34	0.34
SiO <sub>2</sub>	32.97	31.78	31.15	31.15
TiO <sub>2</sub>	0.05	0.05	0.06	0.06
ZnO	0.08	0.08	0.08	0.09
ZrO <sub>2</sub>	0.49	0.50	0.45	0.45

#### Table 3.4c. XRF Analysis (wt%) of Group 3 Formulation Feeds Fused at 1150°C and After VGF Experiments.

Glass Name	e HWI-Al-13		HWI-A	Al-17	HWI-Al-20		
Feed Name	e F-HWI-AL-13-30M F-HWI-AL-13-DHG		F-HWI-AL-17-30M	F-HWI-AL-17-DHG	F-HWI-AL-9KSM-30M	F-HWI-AL-9KSM-DHG	
Sample Condition	VGF (30 minutes)	Fused at 1150°C	VGF (30 minutes)	Fused at 1150°C	VGF (30 minutes)	Fused at 1150°C	
$Al_2O_3$	22.96	22.90	21.66	22.24	23.11	23.23	
$B_2O_3^*$	15.19	15.19	14.73	14.73	18.19	18.19	
BaO	0.07	0.06	0.07	0.06	0.06	0.06	
Bi <sub>2</sub> O <sub>3</sub>	1.19	1.25	1.09	1.24	1.19	1.18	
CaO	6.17	6.31	5.89	6.09	1.08	1.07	
$Cr_2O_3$	0.65	0.67	0.60	0.62	0.63	0.61	
Fe <sub>2</sub> O <sub>3</sub>	5.63	6.03	5.46	5.93	5.87	5.84	
K <sub>2</sub> O	4.02	4.09	2.98	3.02	5.07	5.06	
$Li_2O^*$	3.57	3.57	3.46	3.46	3.17	3.17	
MgO	0.26	0.28	0.33	0.31	0.29	0.27	
Na <sub>2</sub> O	6.10	6.24	9.81	9.60	8.83	9.05	
NiO	0.38	0.41	0.37	0.40	0.37	0.37	
P <sub>2</sub> O <sub>5</sub>	1.28	1.13	1.18	1.13	1.10	1.08	
PbO	0.41	0.45	0.37	0.43	0.33	0.33	
SO <sub>3</sub>	0.33	0.25	0.36	0.18	0.32	0.29	
SiO <sub>2</sub>	31.13	30.48	31.13	29.98	29.86	29.63	
TiO <sub>2</sub>	0.06	0.05	0.05	0.05	0.04	0.04	
ZnO	0.09	0.10	0.08	0.09	0.08	0.08	
ZrO <sub>2</sub>	0.47	0.50	0.35	0.39	0.37	0.38	

#### Table 3.4d. XRF Analysis (wt%) of Group 4 Formulation Feeds Fused at 1150°C and After VGF Experiments.

Glass Name	HWI-A	Al-16	HWI-	Al-16	HWI-A	Al-18
Feed Name	F-HWI-AL-16-30M	F-HWI-AL-16-DHG	F-HWI-AL-16B-30M	F-HWI-AL-16B-DHG	F-HWI-AL-18-30M	F-HWI-AL-18-DHG
Sample Condition	VGF (30 minutes)	Fused at 1150°C	VGF (30 minutes)	Fused at 1150°C	VGF (30 minutes)	Fused at 1150°C
Al <sub>2</sub> O <sub>3</sub>	22.48	22.40	22.55	22.19	21.75	21.39
$B_2O_3^*$	17.73	17.73	17.73	17.73	20.13	20.13
BaO	0.05	0.07	0.05	0.06	0.06	0.06
Bi <sub>2</sub> O <sub>3</sub>	1.16	1.22	1.19	1.20	0.94	0.99
CaO	5.83	5.77	6.07	6.00	5.73	5.86
Cr <sub>2</sub> O <sub>3</sub>	0.60	0.58	0.62	0.62	0.54	0.58
Fe <sub>2</sub> O <sub>3</sub>	5.43	5.38	5.73	5.82	5.60	5.90
K <sub>2</sub> O	0.16	0.14	0.16	0.15	0.17	0.16
Li <sub>2</sub> O*	3.46	3.46	3.46	3.46	3.36	3.36
MgO	0.21	0.25	0.19	0.31	0.28	0.30
Na <sub>2</sub> O	9.87	9.67	9.52	9.00	9.06	9.18
NiO	0.35	0.36	0.37	0.39	0.35	0.39
P <sub>2</sub> O <sub>5</sub>	1.12	1.12	1.11	1.14	1.14	1.13
PbO	0.39	0.64	0.37	0.46	0.35	0.36
SO <sub>3</sub>	0.29	0.22	0.26	0.24	0.20	0.14
SiO <sub>2</sub>	30.23	30.36	30.05	30.64	29.64	29.37
TiO <sub>2</sub>	0.04	0.04	0.03	0.04	0.05	0.04
ZnO	0.09	0.08	0.09	0.10	0.08	0.09
ZrO <sub>2</sub>	0.43	0.43	0.36	0.38	0.46	0.48

#### Table 3.4e. XRF Analysis (wt%) of Group 5 Formulation Feeds Fused at 1150°C and After VGF Experiments.

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Glass Name	HWI-	-Al-19	HWI	Al-19
Feed Name	F-HWI-AL-19- 30M	F-HWI-AL-19- DHG	F-HWI-AL-19B- 30M	F-HWI-AL-19B- DHG
Sample Condition	VGF (30 minutes)	Fused at 1150°C	VGF (30 minutes)	Fused at 1150°C
Al <sub>2</sub> O <sub>3</sub>	23.23	22.91	23.08	22.81
$B_2O_3^*$	19.19	19.19	19.19	19.19
BaO	0.05	0.05	0.05	0.07
Bi <sub>2</sub> O <sub>3</sub>	1.19	1.18	1.14	1.18
CaO	5.62	5.60	5.66	5.71
Cr <sub>2</sub> O <sub>3</sub>	0.63	0.64	0.62	0.62
Fe <sub>2</sub> O <sub>3</sub>	5.80	6.00	5.40	5.63
K <sub>2</sub> O	0.16	0.15	0.16	0.15
$Li_2O^*$	3.57	3.57	3.57	3.57
MgO	0.23	0.33	0.31	0.29
Na <sub>2</sub> O	10.05	10.00	10.47	10.51
NiO	0.38	0.39	0.38	0.38
$P_2O_5$	1.11	1.10	1.09	1.10
PbO	0.29	0.26	0.35	0.38
$SO_3$	0.32	0.17	0.36	0.16
SiO <sub>2</sub>	27.63	27.77	27.67	27.69
TiO <sub>2</sub>	0.05	0.05	0.04	0.04
ZnO	0.09	0.09	0.07	0.08
ZrO <sub>2</sub>	0.36	0.38	0.35	0.39

#### Table 3.4e. XRF Analysis (wt%) of Group 5 Formulation Feeds Fused at 1150°C and After VGF Experiments (continued).

Glass Name	HLW-F	C-Al-27	HLW-F	E-Al-27	HLW-E-Al-27		
Feed Name	F-AL-VAR1-45M	F-AL-VAR1-DHG	F-AL-VAR2-45M	F-AL-VAR2-DHG	F-AL-VAR4-45M	F-AL-VAR4-DHG	
Sample Condition	VGF (45 minutes)	Fused at 1150°C	VGF (45 minutes)	Fused at 1150°C	VGF (45 minutes)	Fused at 1150°C	
Al <sub>2</sub> O <sub>3</sub>	21.59	22.95	23.33	22.97	22.92	23.30	
$B_2O_3^*$	15.19	15.19	15.19	15.19	15.19	15.19	
BaO	0.08	0.08	0.05	0.07	0.06	0.07	
Bi <sub>2</sub> O <sub>3</sub>	1.16	1.19	1.08	1.21	1.15	1.20	
CaO	6.06	6.14	5.93	6.12	5.88	5.93	
Cr <sub>2</sub> O <sub>3</sub>	0.64	0.64	0.62	0.64	0.62	0.63	
Fe <sub>2</sub> O <sub>3</sub>	5.76	5.98	5.54	6.09	5.70	5.90	
K <sub>2</sub> O	0.16	0.15	0.16	0.15	0.15	0.15	
Li <sub>2</sub> O*	3.57	3.57	3.57	3.57	3.57	3.57	
MgO	0.34	0.29	0.33	0.30	0.37	0.36	
MnO	0.00	0.00	0.00	0.00	0.00	0.00	
Na <sub>2</sub> O	9.90	9.42	9.91	10.09	9.75	9.75	
NiO	0.38	0.39	0.37	0.41	0.37	0.38	
$P_2O_5$	1.16	1.04	1.14	1.08	1.13	1.06	
PbO	0.34	0.36	0.27	0.31	0.28	0.29	
SO <sub>3</sub>	0.39	0.29	0.37	0.24	0.35	0.27	
SiO <sub>2</sub>	32.63	31.63	31.54	30.83	31.84	31.23	
TiO <sub>2</sub>	0.04	0.04	0.04	0.04	0.04	0.05	
ZnO	0.07	0.08	0.08	0.09	0.08	0.08	
ZrO <sub>2</sub>	0.48	0.50	0.42	0.51	0.46	0.49	

#### Table 3.4f. XRF Analysis (wt%) of Existing High Fe Melter Feeds and Variations of Baseline Feed for HLW-E-Al-27.

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Glass Name	HLW-E	HLW-E-A1-27		x 1-B1	Matrix 2-9		
Feed Name	BLX-F-50B-60M	BLX-F-50B-DHG	BLS-F-7A	BLS-F-7A	BLV-F-132B	BLV-F-132B	
Sample Condition	VGF (60 minutes)	Fused at 1150°C	VGF (15 minutes)	Fused at 1150°C	VGF (5 minutes)	Fused at 1150°C	
$Al_2O_3$	23.17	23.13	6.04	5.60	12.57	12.50	
$B_2O_3^*$	15.19	15.19	9.41	9.41	13.88	13.88	
BaO	0.06	0.07	0.00	0.00	0.00	0.00	
Bi <sub>2</sub> O <sub>3</sub>	1.09	1.24	0.00	0.00	0.00	0.00	
CaO	5.85	6.12	0.46	0.46	0.31	0.35	
Cr <sub>2</sub> O <sub>3</sub>	0.54	0.61	0.08	0.09	0.08	0.08	
Fe <sub>2</sub> O <sub>3</sub>	5.43	6.02	11.92	12.39	10.30	10.66	
K <sub>2</sub> O	0.18	0.18	0.00	0.00	0.00	0.00	
Li <sub>2</sub> O*	3.57	3.57	3.02	3.02	2.43	2.43	
MgO	0.23	0.23	1.26	1.22	1.26	1.22	
MnO	0.00	0.00	3.37	3.81	2.85	3.19	
Na <sub>2</sub> O	10.60	9.58	12.28	10.78	14.12	12.69	
NiO	0.34	0.38	0.15	0.17	0.13	0.15	
$P_2O_5$	1.20	1.21	0.10	0.10	0.11	0.10	
PbO	0.34	0.37	0.13	0.13	0.11	0.11	
$SO_3$	0.25	0.20	0.00	0.00	0.00	0.00	
SiO <sub>2</sub>	31.38	31.23	46.82	48.05	37.43	38.36	
TiO <sub>2</sub>	0.04	0.04	0.22	0.22	0.23	0.23	
ZnO	0.08	0.09	1.85	1.98	1.55	1.64	
$ZrO_2$	0.34	0.40	0.28	0.31	0.25	0.27	

#### Table 3.4f. XRF Analysis (wt%) of Existing High Fe Melter Feeds and Variations of Baseline Feed for HLW-E-Al-27 (continued).

#### Table 3.5. Ranking Definition for Feed Conversion after 30 Minutes in VGF.

1	Very Fast, all feed converted
2	Fast with minor residue on side wall
3	Moderate with foamy residue on side wall
4	Slow with thick foam layer
5	Slow with partially collapsed dome
6	Very slow with fully developed dome

Al-Limited Was	te Composition	Al-Limited HLW	Al-Limited HLW Waste Simulant			
Waste Oxide	Wt%	Starting Materials	Target Weight (kg)			
$Al_2O_3$	49.21%	Al <sub>2</sub> O <sub>3</sub>	49.707			
B <sub>2</sub> O <sub>3</sub>	0.39%	H <sub>3</sub> BO <sub>3</sub>	0.700			
CaO	2.21%	CaO	2.255			
Fe <sub>2</sub> O <sub>3</sub>	12.11%	Fe(OH) <sub>3</sub> (13% Slurry)	99.643			
Li <sub>2</sub> O	0.35%	Li <sub>2</sub> CO <sub>3</sub>	0.888			
MgO	0.24%	MgO	0.253			
Na <sub>2</sub> O	7.35%	NaOH	4.235			
SiO <sub>2</sub>	10.05%	SiO <sub>2</sub>	10.152			
TiO <sub>2</sub>	0.02%	TiO <sub>2</sub>	0.020			
ZnO	0.17%	ZnO	0.172			
ZrO <sub>2</sub>	0.81%	Zr(OH) <sub>4</sub> ·xH <sub>2</sub> O	2.093			
$SO_3$	0.41%	Na <sub>2</sub> SO <sub>4</sub>	0.735			
Bi <sub>2</sub> O <sub>3</sub>	2.35%	Bi <sub>2</sub> O <sub>3</sub>	2.374			
ThO <sub>2</sub>	0.37%	Om	itted			
Cr <sub>2</sub> O <sub>3</sub>	1.07%	Cr <sub>2</sub> O <sub>3</sub> ·1.5H <sub>2</sub> O	1.273			
K <sub>2</sub> O	0.29%	KNO3	0.632			
$U_3O_8$	7.25%	Om	itted			
BaO	0.11%	BaCO <sub>3</sub>	0.143			
CdO	0.05%	CdO	0.051			
NiO	0.82%	Ni(OH) <sub>2</sub>	1.055			
РЬО	0.84%	PbO	0.848			
$P_2O_5$	2.16%	FePO <sub>4</sub> ·xH <sub>2</sub> O	5.738			
F	1.37%	NaF	3.044			
Carbonate	$1.20^{\#}$	Na <sub>2</sub> CO <sub>3</sub>	0.806			
Nitrite	0.50	NaNO <sub>2</sub>	0.769			
Nitrate	2.00	NaNO <sub>3</sub>	2.230			
Organic Carbon	0.05	$H_2C_2O_4$ ·2 $H_2O$	0.264			
_		Water	279.400			
TOTAL	100.0%	TOTAL	469.478			

Table 3.6. Compositions of the Al-Limited Waste (Oxide Basis) and the HLW Waste Simulant to Produce 100 kg of Waste Oxides (20 wt% suspended solids) (from [2]).

\*Target weights adjusted for assay information of starting materials

<sup>#</sup> Unit for volatile components is g/100 g of waste oxide

Element	Ba	Bi	Cd	Cr	Ni	Pb	Zn
UTS Limits <sup>#</sup>	21	N/A	0.11	0.6	11	0.75	4.3
Delisting Limits	100	N/A	0.48	4.95	22.6	5	225
HLW-E-Al27R1 [2]	0.12	0.91	NA	0.21	0.20	0.37	0.13
HWI-Al-4	0.14	0.55	< 0.03	0.03	< 0.04	0.18	NA
HWI-Al-7	0.17	NA	< 0.03	0.02	0.12	0.43	NA
HWI-Al-16	0.13	NA	< 0.03	0.17	0.27	0.50	NA
HWI-Al-18G	0.17	0.74	0.04	0.08	0.23	0.31	NA
HWI-Al-19	0.17	0.86	< 0.03	0.23	0.23	0.45	NA
HWI-Al-20	0.16	0.78	< 0.03	0.13	< 0.04	0.30	NA

#### Table 3.7. TCLP Results (ppm) for Selected HWI Glasses.

# For comparison only; does not apply to WTP glasses N/A- Not Applicable

NA-Not analyzed

## Table 3.8. Melt Rate Ranking of Feed Samples by VGF vs. Glass Production Rate byDM100 at 1150°C.

Waste Stream	Glass Name	Feed Sample	VGF Melt Rate Ranking*	Glass Production Rate (kg/m²/day)
Al-limited (Al <sub>2</sub> O <sub>3</sub> ) [2]	HLW-E-Al-27	BLL-F-135A	6	550
Al-limited (Al(OH) <sub>3</sub> )	HLW-E-Al-27	BLX-F-50	5	700
Al/Na-limited [2]	HLW-E-ANa-22	BLN-F-21A	3	400
Bi-limited [2]	HLW-E-Bi-6	BLK-F-84A	5	830
Cr-limited [2]	HLW-E-CrM	BLL-F-65A	1	1150
Algorithm Matrix 1 [24]	Matrix 1-B1	BLS-F-7A	3	1000
Algorithm Matrix 2 [25]	Matrix 2-9	BLV-F-132B	3	900
Al-limited (Al <sub>2</sub> O <sub>3</sub> )	HWI-Al-16	F-HWI-Al-16	3	700
Al-limited (Al(OH) <sub>3</sub> )	HWI-Al-16	F-HWI-Al-16B	2	950
Al-limited (Al(OH) <sub>3</sub> )	HWI-Al-19	F-HWI-Al-19B	2	950

\*: Based on Table 3.5

# Table 3.9. Crystalline Phases in the Partially Reacted Feed (VGF) Identified by SEM/EDS and XRD.

Glass Name	Feed Sample (VGF test time)	Al Form in Feed Recipe	Phase identified by SEM/EDS (Surface sample)	Phase identified by XRD (Bulk sample)
3-Al-27	BLL-F-135 (60 minutes)	Al <sub>2</sub> O <sub>3</sub>	<ul> <li>Sodalite (wide spread)</li> <li>Spinel (NiFe)</li> <li>Ca-Phosphate</li> <li>NaCaFeAl-Silicate</li> </ul>	•Quartz •Spinel •Nepheline •Sodalite
HLW-E-Al-27	BLX-F-50 (60 minutes)	Al(OH) <sub>3</sub>	<ul> <li>Sodalite</li> <li>Zr rich oxide with Na,Ca,A, ISi</li> <li>Fe rich oxide with Ca,Al, Si</li> </ul>	•Quartz •Spinel •Nepheline •Sodalite
HWI-Al-16	F-HWI-Al-16 (30 minutes)	Al <sub>2</sub> O <sub>3</sub>	<ul> <li>Spinel (NiCrFe)</li> <li>Ca-Phosphate</li> <li>NaCaFeAl-Silicate</li> <li>CaAl-Silicate with Zr</li> </ul>	•Zircon •Spinel •Sodalite
ΜH	F-HWI-Al-16B (30 minutes)	Al(OH) <sub>3</sub>	<ul><li> Spinel (NiCrFe)</li><li> Zr oxide with Ca, Al, Si</li></ul>	•Zircon •Spinel •Sodalite
Al-19	F-HWI-Al-19F (30 minutes)	Al <sub>2</sub> O <sub>3</sub>	<ul> <li>Sodalite (wide spread)</li> <li>Spinel (NiCrFe)</li> <li>Ca-Phosphate</li> </ul>	•Zircon •Spinel •Sodalite •Quartz
61-IV-IMH F-HWI-Al-19B (30 minutes)		Al(OH) <sub>3</sub>	<ul> <li>NaCaAlFe-Silicate</li> <li>Spinel (NiCrFe)</li> <li>Zr oxide with Ca, Al, Si</li> </ul>	•Zircon •Spinel •Sodalite •Quartz

Base Feed		E-Al-27 with ee Table 2.2			Al-27 with ee Table 2.	· /•	HLW-E-Al-27 with Al <sub>2</sub> O <sub>3</sub> and Boric Acid/Soda Ash Replacing Borax			HLW-E-Al-27 with Al <sub>2</sub> O <sub>3</sub> ,and Boric Acid/ K <sub>2</sub> CO <sub>3</sub> Replacing Borax			
Test s	50, 51, 79, 105, 153	64,	65	66, 92, 118, 149, 205	86, 115, 147		67, 93	87,	116	68, 94, 119, 126, 155	88, 11	7, 148,	
Feed (g)	1000	95	50	1000	950		1000	9:	50	1000	95	50	
Type of Additive	None	HNO <sub>3</sub>	Sugar	None	HNO <sub>3</sub>	Sugar	None	HNO <sub>3</sub>	Sugar	None	HNO <sub>3</sub>	Sugar	
Additive	0 g	50 ml	17g	0	50 ml	17g	0g	50 ml	17g	0g	50 ml	17g	
Al <sub>2</sub> O <sub>3</sub>	23.97	23.	97	23.97	23	.97	23.97	23	.97	23.97	23.	97	
B <sub>2</sub> O <sub>3</sub>	15.19	15.	19	15.19	15	.19	15.19	15	.19	15.19	15.	19	
BaO	0.05	0.0	)5	0.05	0.	05	0.05	0.	05	0.05	0.0	05	
Bi <sub>2</sub> O <sub>3</sub>	1.14	1.	14	1.14	1.14		1.14	1.14		1.14	1.	14	
CaO	6.08	6.0	)8	6.08	6.08		6.08	6.08		6.08	6.08		
CdO	0.02	0.0	)2	0.02	0.02		0.02	0.02		0.02	0.02		
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.:	52	0.52	0.52		0.52	0.	52	0.52	0.:	52	
F	0.67	0.0	57	0.67	0.67		0.67	0.	67	0.67	0.0	57	
Fe <sub>2</sub> O <sub>3</sub>	5.90	5.9	90	5.90	5.	90	5.90	5.	90	5.90	5.90		
K <sub>2</sub> O	0.14	0.	14	0.14	0.	14	0.14	0.14		6.14	6.14		
Li <sub>2</sub> O	3.57	3.:	57	3.57	3.	57	3.57	3.57		3.57	3.57		
MgO	0.12	0.	12	0.12	0.	12	0.12	0.12		0.12	0.12		
Na <sub>2</sub> O	9.58	9.:	58	9.58	9.	58	9.58	9.	58	3.58	3.:	58	
NiO	0.40	0.4	40	0.40	0.	40	0.40	0.	40	0.40	0.4	40	
$P_2O_5$	1.05	1.0	)5	1.05	1.	05	1.05	1.	05	1.05	1.0	05	
PbO	0.41	0.4	41	0.41	0.	41	0.41	0.	41	0.41	0.4	41	
SiO <sub>2</sub>	30.50	30.	50	30.50	30	.50	30.50	30	.50	30.50	30.	50	
SO <sub>3</sub>	0.20	0.2	20	0.20	0.	20	0.20	0.	20	0.20	0.2	20	
TiO <sub>2</sub>	0.01	0.0	)1	0.01	0.	01	0.01	0.	01	0.01	0.0	01	
ZnO	0.08	0.0	)8	0.08	0.	08	0.08	0.08		0.08	0.08		
ZrO <sub>2</sub>	0.39	0.3	39	0.39	0.	39	0.39	0.39		0.39	0.39		
Sum	100.00	100	.00	100.00	100	0.00	100.00	100	0.00	100.00	100.00		

#### Table 3.10. Target Glass Compositions for Base Feeds Used During DM10 Melter Rate Tests (wt%).

Melt Rate Enhancement for High Aluminum HLW Glass Formulations Final Report, VSL-07R1010-1, Rev. 0

Feed type		Н	WI-Al-9Me	lt		HWI-Al-9KSM					
Test #s	154, 166, 180	173,	202	175,	204	156, 179, 181	174,	203	17	76	
Feed (g)	1000	95	50	975		1000	950		975		
Type of Additive	None	HNO3	Sugar	HNO3	Sugar	None	HNO3	Sugar	HNO3	Sugar	
Additive	0 g	50 ml	17g	25 ml	8.5g	0g	50 ml	17g	25 ml	8.5g	
Al <sub>2</sub> O <sub>3</sub>	21.31	21.	31	21.	31	23.97	23.	97	23.	97	
B <sub>2</sub> O <sub>3</sub>	18.17	18.	17	18.	17	18.19	18.	19	18	19	
BaO	0.05	0.0	05	0.0	)5	0.05	0.0	)5	0.0	)5	
Bi <sub>2</sub> O <sub>3</sub>	1.02	1.0	02	1.0	02	1.14	1.1	14	1.	14	
CaO	0.96	0.9	96	0.9	96	1.08	1.0	08	1.0	)8	
CdO	0.02	0.0	02	0.02		0.02	0.02		0.	)2	
Cr <sub>2</sub> O <sub>3</sub>	0.46	0.4	46	0.46		0.52	0.52		0.:	52	
F	0.59	0.59		0.59		0.67	0.0	57	0.	57	
Fe <sub>2</sub> O <sub>3</sub>	5.24	5.24		5.24		5.90	5.9	90	5.	90	
K <sub>2</sub> O	0.13	0.	13	0.13		5.14	5.	14	5.	14	
Li <sub>2</sub> O	3.15	3.	15	3.15		3.17	3.17		3.	17	
MgO	0.10	0.	10	0.	0.10		0.12		0.	12	
Na <sub>2</sub> O	13.18	13.	18	13.	18	8.58	8.58		8.58		
NiO	0.36	0.1	36	0	36	0.40	0.40		0.4	40	
P <sub>2</sub> O <sub>5</sub>	0.94	0.9	94	0.9	94	1.05	1.05		1.0	)5	
PbO	0.36	0.1	36	0.1	36	0.41	0.41		0.4	41	
SiO <sub>2</sub>	33.35	33.	35	33.	35	28.90	28.	90	28.	90	
SO <sub>3</sub>	0.18	0.	18	0.	18	0.20	0.2	20	0.2	20	
TiO <sub>2</sub>	0.01	0.0	01	0.0	)1	0.01	0.0	)1	0.0	)1	
ZnO	0.07	0.0	07	0.0	)7	0.08	0.08		0.0	)8	
ZrO <sub>2</sub>	0.35	0.3	35	0.1	35	0.39	0.39		0.	39	
Sum	100.00	100	.00	100	100.00		100.00		100.00		

#### Table 3.10. Target Glass Compositions for Base Feeds Used During DM10 Melter Rate Tests (wt%) (continued).

Melt Rate Enhancement for High Aluminum HLW Glass Formulations Final Report, VSL-07R1010-1, Rev. 0

Base Feed					HL	W-E-Al-27	with Al <sub>2</sub> C	)3				
Test #s	50, 51, 64, 65, 79, 105, 153	58, 59	52, 53	75, 101, 124	56, 57	60, 61	54, 55	62, 63	159, 182	199	162, 186	167, 192
Feed (g)	950-1000	950	950	950	950	950	950	950	900	900	900	900
Type of Additive	None	Boric Acid	Na2CO3	K2CO3	Borax	Li2CO3	CaCO3	SiO2	Boric Acid	Na2CO3	K2CO3	Borax
Additive (g)	0	50	50	50	50	50	50	50	100	100	100	100
Al <sub>2</sub> O <sub>3</sub>	23.97	22.17	22.12	21.85	21.82	22.66	22.19	20.98	20.47	20.36	19.89	19.84
$B_2O_3$	15.19	21.55	14.01	13.84	20.04	14.36	14.06	13.29	27.58	12.90	12.60	24.50
BaO	0.05	0.05	0.05	0.05	0.05	0.05	0.05	0.05	0.05	0.05	0.04	0.04
Bi <sub>2</sub> O <sub>3</sub>	1.14	1.06	1.06	1.04	1.04	1.08	1.06	1.00	0.98	0.97	0.95	0.95
CaO	6.08	5.62	5.61	5.54	5.53	5.74	13.06	5.32	5.19	5.16	5.04	5.03
CdO	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.48	0.48	0.47	0.47	0.49	0.48	0.46	0.44	0.44	0.43	0.43
F	0.67	0.62	0.62	0.61	0.61	0.63	0.62	0.58	0.57	0.57	0.55	0.55
Fe <sub>2</sub> O <sub>3</sub>	5.90	5.46	5.44	5.38	5.37	5.58	5.46	5.16	5.04	5.01	4.89	4.88
K <sub>2</sub> O	0.14	0.13	0.13	9.00	0.13	0.13	0.13	0.12	0.12	0.12	17.16	0.12
Li <sub>2</sub> O	3.57	3.30	3.29	3.25	3.25	8.83	3.30	3.12	3.05	3.03	2.96	2.96
MgO	0.12	0.11	0.11	0.11	0.11	0.11	0.11	0.10	0.10	0.10	0.10	0.10
Na <sub>2</sub> O	9.58	8.86	16.58	8.73	11.49	9.06	8.87	8.38	8.18	23.19	7.95	13.24
NiO	0.40	0.37	0.37	0.36	0.36	0.38	0.37	0.35	0.34	0.34	0.33	0.33
P <sub>2</sub> O <sub>5</sub>	1.05	0.97	0.97	0.96	0.96	0.99	0.97	0.92	0.90	0.89	0.87	0.87
PbO	0.41	0.38	0.38	0.37	0.37	0.39	0.38	0.36	0.35	0.35	0.34	0.34
SiO <sub>2</sub>	30.50	28.21	28.13	27.79	27.76	28.83	28.23	39.18	26.04	25.91	25.30	25.24
SO <sub>3</sub>	0.20	0.18	0.18	0.18	0.18	0.19	0.18	0.17	0.17	0.17	0.17	0.17
TiO <sub>2</sub>	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01
ZnO	0.08	0.08	0.08	0.08	0.08	0.08	0.08	0.07	0.07	0.07	0.07	0.07
ZrO <sub>2</sub>	0.39	0.37	0.36	0.36	0.36	0.37	0.37	0.35	0.34	0.34	0.33	0.33
Sum	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00

#### Table 3.11. Target Compositions for Vitrified Melter Feed with Various Additives During DM10 Melter Rate Tests (wt%).

Melt Rate Enhancement for High Aluminum HLW Glass Formulations Final Report, VSL-07R1010-1, Rev. 0

Base Feed	HLW-E-Al-27 with Al <sub>2</sub> O <sub>3</sub>												
Test #s	50, 51, 79, 105, 153	160, 183	200	163, 188, 208	168, 193	161, 184	164, 189	194	177, 185	190	196		
Feed (g)	1000	850	850	850	850	800	800	800	750	750	750		
Type of Additive	None	Boric Acid	Na2CO3	K2CO3	Borax	Boric Acid	K2CO3	Borax	Boric Acid	K2CO3	Borax		
Additives (g)	0	150	150	150	150	200	200	200	250	250	250		
$Al_2O_3$	23.97	18.85	18.71	18.07	18.01	17.31	16.39	16.32	15.84	14.83	14.76		
$B_2O_3$	15.19	33.31	11.85	11.45	28.61	38.76	10.39	32.42	43.95	9.40	35.95		
BaO	0.05	0.04	0.04	0.04	0.04	0.04	0.04	0.04	0.04	0.03	0.03		
Bi <sub>2</sub> O <sub>3</sub>	1.14	0.90	0.89	0.86	0.86	0.83	0.78	0.78	0.76	0.71	0.70		
CaO	6.08	4.78	4.74	4.58	4.57	4.39	4.16	4.14	4.02	3.76	3.74		
CdO	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.01		
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.41	0.41	0.39	0.39	0.38	0.36	0.35	0.34	0.32	0.32		
F	0.67	0.52	0.52	0.50	0.50	0.48	0.46	0.45	0.44	0.41	0.41		
Fe <sub>2</sub> O <sub>3</sub>	5.90	4.64	4.60	4.45	4.43	4.26	4.03	4.02	3.90	3.65	3.63		
K <sub>2</sub> O	0.14	0.11	0.11	24.70	0.11	0.10	31.70	0.10	0.09	38.21	0.09		
Li <sub>2</sub> O	3.57	2.81	2.79	2.69	2.68	2.58	2.44	2.43	2.36	2.21	2.20		
MgO	0.12	0.09	0.09	0.09	0.09	0.08	0.08	0.08	0.08	0.07	0.07		
Na <sub>2</sub> O	9.58	7.53	29.43	7.22	14.85	6.92	6.55	16.35	6.33	5.93	17.74		
NiO	0.40	0.31	0.31	0.30	0.30	0.29	0.27	0.27	0.26	0.25	0.25		
$P_2O_5$	1.05	0.83	0.82	0.79	0.79	0.76	0.72	0.72	0.70	0.65	0.65		
PbO	0.41	0.32	0.32	0.31	0.31	0.30	0.28	0.28	0.27	0.25	0.25		
SiO <sub>2</sub>	30.50	23.98	23.80	22.99	22.92	22.02	20.86	20.77	20.15	18.87	18.77		
SO <sub>3</sub>	0.20	0.16	0.16	0.15	0.15	0.14	0.14	0.14	0.13	0.12	0.12		
TiO <sub>2</sub>	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01		
ZnO	0.08	0.07	0.06	0.06	0.06	0.06	0.06	0.06	0.05	0.05	0.05		
ZrO <sub>2</sub>	0.39	0.31	0.31	0.30	0.30	0.28	0.27	0.27	0.26	0.24	0.24		
Sum	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00		

### Table 3.11. Target Compositions for Vitrified Melter Feed with Various Additives During DM10 Melter Rate Tests (wt%)(continued).

Melt Rate Enhancement for High Aluminum HLW Glass Formulations Final Report, VSL-07R1010-1, Rev. 0

Base Feed				HLW-	E-Al-27 with	n Al(OH)3				
Test #s	66, 86, 92, 115, 118, 147, 149, 205	69, 95, 120	72, 98, 122	76, 102, 125	80, 106, 145	83, 109	89, 112	207, 209	211, 213	222, 226
Feed (g)	950-1000	950	950	950	950	950	950	900	900	900
Type of Additive	None	Boric Acid	Na2CO3	K2CO3	Borax	Li2CO3	CaCO3	Boric Acid	K2CO3	Borax
Additive (g)	0	50	50	50	50	50	50	100	100	100
$Al_2O_3$	23.97	22.17	22.12	21.85	21.82	22.66	22.19	20.47	19.89	19.84
$B_2O_3$	15.19	21.55	14.01	13.84	20.04	14.36	14.06	27.58	12.60	24.50
BaO	0.05	0.05	0.05	0.05	0.05	0.05	0.05	0.05	0.04	0.04
Bi <sub>2</sub> O <sub>3</sub>	1.14	1.06	1.06	1.04	1.04	1.08	1.06	0.98	0.95	0.95
CaO	6.08	5.62	5.61	5.54	5.53	5.74	13.06	5.19	5.04	5.03
CdO	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.48	0.48	0.47	0.47	0.49	0.48	0.44	0.43	0.43
F	0.67	0.62	0.62	0.61	0.61	0.63	0.62	0.57	0.55	0.55
Fe <sub>2</sub> O <sub>3</sub>	5.90	5.46	5.44	5.38	5.37	5.58	5.46	5.04	4.89	4.88
K <sub>2</sub> O	0.14	0.13	0.13	9.00	0.13	0.13	0.13	0.12	17.16	0.12
Li <sub>2</sub> O	3.57	3.30	3.29	3.25	3.25	8.83	3.30	3.05	2.96	2.96
MgO	0.12	0.11	0.11	0.11	0.11	0.11	0.11	0.10	0.10	0.10
Na <sub>2</sub> O	9.58	8.86	16.58	8.73	11.49	9.06	8.87	8.18	7.95	13.24
NiO	0.40	0.37	0.37	0.36	0.36	0.38	0.37	0.34	0.33	0.33
$P_2O_5$	1.05	0.97	0.97	0.96	0.96	0.99	0.97	0.90	0.87	0.87
PbO	0.41	0.38	0.38	0.37	0.37	0.39	0.38	0.35	0.34	0.34
SiO <sub>2</sub>	30.50	28.21	28.13	27.79	27.76	28.83	28.23	26.04	25.30	25.24
SO <sub>3</sub>	0.20	0.18	0.18	0.18	0.18	0.19	0.18	0.17	0.17	0.17
TiO <sub>2</sub>	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01
ZnO	0.08	0.08	0.08	0.08	0.08	0.08	0.08	0.07	0.07	0.07
ZrO <sub>2</sub>	0.39	0.37	0.36	0.36	0.36	0.37	0.37	0.34	0.33	0.33
Sum	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00

### Table 3.11. Target Compositions for Vitrified Melter Feed with Various Additives During DM10 Melter Rate Tests (wt%) (continued).

Base Feed	HLW-E-Al-27 with Al(OH) <sub>3</sub>												
Test #s	66, 92, 118, 149, 205	210	200	212	223, 227	214, 216	215, 217	224, 228	218, 220	219, 221	225		
Feed (g)	1000	850	850	850	850	800	800	800	750	750	750		
Type of Additive	None	Boric Acid	Na2CO3	K2CO3	Borax	Boric Acid	K2CO3	Borax	Boric Acid	K2CO3	Borax		
Additives (g)	0	150	150	150	150	200	200	200	250	250	250		
Al <sub>2</sub> O <sub>3</sub>	23.97	18.85	18.71	18.07	18.01	17.31	16.39	16.32	15.84	14.83	14.76		
$B_2O_3$	15.19	33.31	11.85	11.45	28.61	38.76	10.39	32.42	43.95	9.40	35.95		
BaO	0.05	0.04	0.04	0.04	0.04	0.04	0.04	0.04	0.04	0.03	0.03		
Bi <sub>2</sub> O <sub>3</sub>	1.14	0.90	0.89	0.86	0.86	0.83	0.78	0.78	0.76	0.71	0.70		
CaO	6.08	4.78	4.74	4.58	4.57	4.39	4.16	4.14	4.02	3.76	3.74		
CdO	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.01		
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.41	0.41	0.39	0.39	0.38	0.36	0.35	0.34	0.32	0.32		
F	0.67	0.52	0.52	0.50	0.50	0.48	0.46	0.45	0.44	0.41	0.41		
Fe <sub>2</sub> O <sub>3</sub>	5.90	4.64	4.60	4.45	4.43	4.26	4.03	4.02	3.90	3.65	3.63		
K <sub>2</sub> O	0.14	0.11	0.11	24.70	0.11	0.10	31.70	0.10	0.09	38.21	0.09		
Li <sub>2</sub> O	3.57	2.81	2.79	2.69	2.68	2.58	2.44	2.43	2.36	2.21	2.20		
MgO	0.12	0.09	0.09	0.09	0.09	0.08	0.08	0.08	0.08	0.07	0.07		
Na <sub>2</sub> O	9.58	7.53	29.43	7.22	14.85	6.92	6.55	16.35	6.33	5.93	17.74		
NiO	0.40	0.31	0.31	0.30	0.30	0.29	0.27	0.27	0.26	0.25	0.25		
$P_2O_5$	1.05	0.83	0.82	0.79	0.79	0.76	0.72	0.72	0.70	0.65	0.65		
PbO	0.41	0.32	0.32	0.31	0.31	0.30	0.28	0.28	0.27	0.25	0.25		
SiO <sub>2</sub>	30.50	23.98	23.80	22.99	22.92	22.02	20.86	20.77	20.15	18.87	18.77		
$SO_3$	0.20	0.16	0.16	0.15	0.15	0.14	0.14	0.14	0.13	0.12	0.12		
TiO <sub>2</sub>	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01		
ZnO	0.08	0.07	0.06	0.06	0.06	0.06	0.06	0.06	0.05	0.05	0.05		
ZrO <sub>2</sub>	0.39	0.31	0.31	0.30	0.30	0.28	0.27	0.27	0.26	0.24	0.24		
Sum	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00		

## Table 3.11. Target Compositions for Vitrified Melter Feed with Various Additives During DM10 Melter Rate Tests (wt%),<br/>(continued).

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The Catholic University of America Vitreous State Laboratory

ZnO

ZrO<sub>2</sub>

Sum

0.08

0.39

100.00

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0.08

0.39

100.00

0.08

0.36

100.00

0.08

0.36

100.00

0.07

0.36

100.00

0.07

0.36

100.00

Base Feed	HLW-E-	-Al-27 with	$Al_2O_3$ and	Boric Acid/	'Soda Ash R	Replacing B	orax	HLW-E-Al-27 wit	h Al <sub>2</sub> O <sub>3</sub> , and	d Boric Aci	d/ K <sub>2</sub> CO <sub>3</sub> I	Replacing		
							0	Borax						
Test #s	67, 87, 93, 116	70, 96	73, 99	77, 103	81, 107	84, 110	90, 113	68, 88, 94, 117, 119, 126, 148, 155	71, 97, 121	74, 100, 123	78, 104	82, 108		
Feed (g)	950-1000	950	950	950	950	950	950	950-1000	950	950	950	950		
Type of Additive	None	Boric Acid	Na2CO3	K2CO3	Borax	Li2CO3	CaCO3	None	Boric Acid	Na2CO3	K2CO3	Borax		
Additives (g)	0	50	50	50	50	50	50	0	50	50	50	50		
$Al_2O_3$	23.97	22.17	22.12	21.85	21.81	22.66	22.19	23.97	22.03	21.97	21.67	21.63		
$B_2O_3$	15.19	21.55	14.01	13.84	20.08	14.36	14.06	15.19	22.07	13.92	13.73	20.47		
BaO	0.05	0.05	0.05	0.05	0.05	0.05	0.05	0.05	0.05	0.05	0.05	0.05		
Bi <sub>2</sub> O <sub>3</sub>	1.14	1.06	1.06	1.04	1.04	1.08	1.06	1.14	1.05	1.05	1.04	1.03		
CaO	6.08	5.62	5.61	5.54	5.53	5.74	13.07	6.08	5.58	5.57	5.49	5.48		
CdO	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02		
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.48	0.48	0.47	0.47	0.49	0.48	0.52	0.48	0.48	0.47	0.47		
F	0.67	0.62	0.62	0.61	0.61	0.63	0.62	0.67	0.61	0.61	0.60	0.60		
Fe <sub>2</sub> O <sub>3</sub>	5.90	5.46	5.44	5.38	5.37	5.58	5.46	5.90	5.42	5.41	5.33	5.32		
K <sub>2</sub> O	0.14	0.13	0.13	9.00	0.13	0.13	0.13	6.14	5.64	5.63	15.14	5.54		
Li <sub>2</sub> O	3.57	3.30	3.29	3.25	3.25	8.83	3.30	3.57	3.28	3.27	3.23	3.22		
MgO	0.12	0.11	0.11	0.11	0.11	0.11	0.11	0.12	0.11	0.11	0.11	0.11		
Na <sub>2</sub> O	9.58	8.86	16.58	8.73	11.48	9.06	8.87	3.58	3.29	11.62	3.24	6.22		
NiO	0.40	0.37	0.37	0.36	0.36	0.38	0.37	0.40	0.37	0.37	0.36	0.36		
$P_2O_5$	1.05	0.97	0.97	0.96	0.96	0.99	0.97	1.05	0.97	0.96	0.95	0.95		
PbO	0.41	0.38	0.38	0.37	0.37	0.39	0.38	0.41	0.38	0.37	0.37	0.37		
SiO <sub>2</sub>	30.50	28.21	28.13	27.79	27.74	28.83	28.23	30.50	28.02	27.95	27.57	27.52		
$SO_3$	0.20	0.18	0.18	0.18	0.18	0.19	0.18	0.20	0.18	0.18	0.18	0.18		
TiO <sub>2</sub>	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01		

## Table 3.11. Target Compositions for Vitrified Melter Feed with Various Additives During DM10 Melter Rate Tests (wt%) (continued).

0.08

0.37

100.00

0.08

0.37

100.00

0.08

0.36

100.00

0.08

0.36

100.00

0.08

0.36

100.00

0.08

0.36

100.00

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Table 3.11. Target Compositions for Vitrified Melter Feed with Various Additives During DM10 Melter Rate Tests (wt%)
(continued).

Base Feed	HLW-	E-Al-27 w	vith Al <sub>2</sub> O	3 ,and Boric	Acid/ K <sub>2</sub>	CO <sub>3</sub> Repl	acing Bo	orax	HWI		HWI-Al-9KSM			
Test #s	68, 94, 119, 126, 155	85, 111	91, 114	178	165	169, 178	187	172	154, 166, 173, 175, 180, 202, 204	158,	170	156, 174, 176, 179, 181, 203	157, 206	171
Feed (g)	1000	950	950	900	900	900	850	850	950-1000	950	950	950-1000	950	950
Type of Additive	None	Li2CO3	CaCO3	Boric Acid	K2CO3	Borax	K2CO3	Borax	None	Boric Acid	Borax	None	Boric Acid	Borax
Additives (g)	0	50	50	100	100	100	150	150	0	50	50	0	50	50
Al <sub>2</sub> O <sub>3</sub>	23.97	22.55	22.04	20.21	19.59	19.52	17.68	17.60	21.31	19.70	19.38	23.97	22.17	21.82
$B_2O_3$	15.19	14.29	13.97	28.51	12.41	25.25	11.21	29.60	18.17	24.34	22.78	18.19	24.32	22.77
BaO	0.05	0.05	0.05	0.05	0.04	0.04	0.04	0.04	0.05	0.04	0.04	0.05	0.05	0.05
Bi <sub>2</sub> O <sub>3</sub>	1.14	1.08	1.05	0.96	0.94	0.93	0.84	0.84	1.02	0.94	0.93	1.14	1.06	1.04
CaO	6.08	5.72	13.64	5.12	4.97	4.95	4.48	4.46	0.96	0.88	0.87	1.08	1.00	0.98
CdO	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02	0.02
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.49	0.48	0.44	0.43	0.42	0.38	0.38	0.46	0.43	0.42	0.52	0.48	0.47
F	0.67	0.63	0.61	0.56	0.55	0.54	0.49	0.49	0.59	0.55	0.54	0.67	0.62	0.61
Fe <sub>2</sub> O <sub>3</sub>	5.90	5.55	5.42	4.97	4.82	4.80	4.35	4.33	5.24	4.85	4.77	5.90	5.46	5.37
K <sub>2</sub> O	6.14	5.78	5.65	5.18	23.31	5.00	30.75	4.51	0.13	0.12	0.11	5.14	4.76	4.68
Li <sub>2</sub> O	3.57	9.28	3.28	3.01	2.92	2.91	2.63	2.62	3.15	2.91	2.87	3.17	2.93	2.89
MgO	0.12	0.11	0.11	0.10	0.10	0.10	0.09	0.09	0.10	0.10	0.09	0.12	0.11	0.11
Na <sub>2</sub> O	3.58	3.37	3.29	3.02	2.93	8.60	2.64	10.77	13.18	12.19	14.77	8.58	7.94	10.57
NiO	0.40	0.38	0.37	0.34	0.33	0.33	0.29	0.29	0.36	0.33	0.32	0.40	0.37	0.36
$P_2O_5$	1.05	0.99	0.97	0.89	0.86	0.86	0.78	0.77	0.94	0.86	0.85	1.05	0.97	0.96
PbO	0.41	0.38	0.38	0.34	0.33	0.33	0.30	0.30	0.36	0.34	0.33	0.41	0.38	0.37
SiO <sub>2</sub>	30.50	28.69	28.04	25.71	24.92	24.83	22.50	22.39	33.35	30.84	30.34	28.90	26.73	26.30
SO3	0.20	0.19	0.18	0.17	0.16	0.16	0.15	0.15	0.18	0.16	0.16	0.20	0.18	0.18
TiO <sub>2</sub>	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01
ZnO	0.08	0.08	0.08	0.07	0.07	0.07	0.06	0.06	0.07	0.07	0.07	0.08	0.08	0.08
ZrO <sub>2</sub>	0.39	0.37	0.36	0.33	0.32	0.32	0.29	0.29	0.35	0.32	0.32	0.39	0.36	0.36
Sum	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00

Table 3.12. Composition and Properties of Aluminum Limited Waste and Glass	
Formulation HWI-Al-16 with 43.65% Waste Loading (wt%).	

-	Al-Limited Waste <sup>*</sup>	Waste in Glass	Glass Forming Additives	Target Glass HWI-Al-16
Al <sub>2</sub> O <sub>3</sub>	53.27	23.25	-	23.25
$B_2O_3$	0.42	0.18	17.55	17.73
BaO	0.12	0.05	-	0.05
Bi <sub>2</sub> O <sub>3</sub>	2.54	1.11	-	1.11
CaO	2.39	1.04	4.85	5.89
CdO	0.05	0.02	-	0.02
Cr <sub>2</sub> O <sub>3</sub>	1.16	0.51		0.51
F	1.48	0.65	-	0.65
Fe <sub>2</sub> O <sub>3</sub>	13.11	5.72	-	5.72
K <sub>2</sub> O	0.31	0.14	-	0.14
Li <sub>2</sub> O	0.38	0.17	3.30	3.46
MgO	0.26	0.11	-	0.11
Na <sub>2</sub> O	7.96	3.47	5.82	9.29
NiO	0.89	0.39	-	0.39
$P_2O_5$	2.34	1.02	-	1.02
PbO	0.91	0.40	-	0.40
$SO_3$	0.44	0.19	-	0.19
SiO <sub>2</sub>	10.88	4.75	24.83	29.58
TiO <sub>2</sub>	0.02	0.01	-	0.01
ZnO	0.18	0.08	-	0.08
ZrO <sub>2</sub>	0.88	0.38	-	0.38
Sum	100.0	43.65	56.35	100.0#

\* Renormalized from Ref. [5] after removal of radioactive components. # The sum does not equal to 100.00 because of rounding of decimals.

Viscosi	43		
Conductivi	0.24		
Crystal Co	None		
Crystal Con	1.6%		
Crystal	1.9%		
	Pass		
	-	DWPF-EA	HWI-Al-16
PCT, g/L	В	16.7	0.314
1 C 1, g/L	Li	9.6	0.394
	Na	13.3	0.285

-	Al-Limited Waste <sup>*</sup>	Waste in Glass	Glass Forming Additives	Target Glass HWI-Al-19
Al <sub>2</sub> O <sub>3</sub>	53.27	23.97	-	23.97
$B_2O_3$	0.42	0.19	19.00	19.19
BaO	0.12	0.05	-	0.05
Bi <sub>2</sub> O <sub>3</sub>	2.54	1.14	-	1.14
CaO	2.39	1.08	4.50	5.58
CdO	0.05	0.02	-	0.02
Cr <sub>2</sub> O <sub>3</sub>	1.16	0.52		0.52
F	1.48	0.67	-	0.67
Fe <sub>2</sub> O <sub>3</sub>	13.11	5.90	-	5.90
K <sub>2</sub> O	0.31	0.14	-	0.14
Li <sub>2</sub> O	0.38	0.17	3.40	3.57
MgO	0.26	0.12	-	0.12
Na <sub>2</sub> O	7.96	3.58	6.00	9.58
NiO	0.89	0.40	-	0.40
$P_2O_5$	2.34	1.05	-	1.05
PbO	0.91	0.41	-	0.41
$SO_3$	0.44	0.20	-	0.20
SiO <sub>2</sub>	10.88	4.90	22.10	27.00
TiO <sub>2</sub>	0.02	0.01	-	0.01
ZnO	0.18	0.08	-	0.08
ZrO <sub>2</sub>	0.88	0.39	-	0.39
Sum	100.0	45.00	55.00	$100.0^{\#}$

#### Table 3.13. Composition and Properties of Aluminum Limited Waste and Glass Formulation HWI-Al-19 with 45% Waste Loading (wt%).

\* Renormalized from Ref. [5] after removal of radioactive components. # The sum does not equal to 100.00 because of rounding of decimals.

Viscosi	Viscosity @1150°C, P					
Conductivi	0.27					
Crystal Co	ontent, As Melted	None				
Crystal Con	nt, 72 hr at 950°C 1.3					
Crystal	Content, CCC		1.9			
	TCLP		Pass			
	-	DWPF-EA	HWI-Al-19			
PCT, g/L	В	, S/cm elted 950°C CC DWPF-EA 16.7 9.6	0.654			
1 C 1, g/L	Li	0.794				
	Na	13.3	0.624			

- Empty data field

#### Table 3.14. Composition of Melter Feed to Produce 100 kg of Target Glass HWI-Al-16 (Target Glass Yield = 500 g/L Feed) from the Al-Limited Waste Simulant Using Al(OH)<sub>3</sub> as the Aluminum Source.

Al-Limited W	aste Simulant	Glass-Formi	ng Additives
Starting Materials	Target Weight (kg) <sup>*</sup>	Starting Materials	Target Weight (kg) *
Al(OH) <sub>3</sub>	35.935	_	
H <sub>3</sub> BO <sub>3</sub>	0.331	H <sub>3</sub> BO <sub>3</sub>	31.487
BaCO <sub>3</sub>	0.068	—	
Bi <sub>2</sub> O <sub>3</sub>	1.122	—	
CaO	1.066	CaSiO <sub>3</sub> (Wollastonite)	10.560
CdO	0.024	_	
Cr <sub>2</sub> O <sub>3</sub>	0.516	—	
NaF	1.438	—	
Fe(OH) <sub>3</sub> (13% Slurry)	47.083	_	
KNO3	0.299	—	
Li <sub>2</sub> CO <sub>3</sub>	0.420	Li <sub>2</sub> CO <sub>3</sub>	8.366
MgO	0.117	—	
NaOH	2.124	Na <sub>2</sub> CO <sub>3</sub>	10.053
Ni(OH) <sub>2</sub>	0.498	—	
FePO <sub>4</sub> ·xH <sub>2</sub> O	2.711	—	
PbO	0.401	_	
$Na_2SO_4$	0.347	—	
SiO <sub>2</sub>	4.797	SiO <sub>2</sub>	19.643
TiO <sub>2</sub>	0.010	_	
ZnO	0.081	—	
Zr(OH) <sub>4</sub> ·xH <sub>2</sub> O	0.989	_	
H <sub>2</sub> O	95.874	—	
Na <sub>2</sub> CO <sub>3</sub>	0.304	—	
NaNO <sub>2</sub>	0.336		
NaNO <sub>3</sub>	0.954		
$H_2C_2O_4$ ·2 $H_2O$	0.115		
		—	
Simulant Total	197.960	Additives Total	80.109
		FEED TOTAL	278.069

\* Target weights adjusted for assay information of starting materials — Empty data field

# Table 3.15. Composition of Melter Feed to Produce 100 kg of Target Glass HWI-Al-16(Target Glass Yield = 500 g/L Feed) from the Al-Limited Waste Simulant Using Al<sub>2</sub>O<sub>3</sub> as<br/>the Aluminum Source.

Al-Limited W	aste Simulant	Glass-Formi	ng Additives
Starting Materials	Target Weight (kg) <sup>*</sup>	Starting Materials	Target Weight (kg)*
Al <sub>2</sub> O <sub>3</sub>	23.487	_	
H <sub>3</sub> BO <sub>3</sub>	0.331	H <sub>3</sub> BO <sub>3</sub>	31.487
BaCO <sub>3</sub>	0.068	—	
Bi <sub>2</sub> O <sub>3</sub>	1.122	—	
CaO	1.066	CaSiO <sub>3</sub> (Wollastonite)	10.560
CdO	0.024	_	—
Cr <sub>2</sub> O <sub>3</sub>	0.516	—	
NaF	1.438	—	
Fe(OH) <sub>3</sub> (13% Slurry)	47.083	—	_
KNO3	0.299	—	—
Li <sub>2</sub> CO <sub>3</sub>	0.420	Li <sub>2</sub> CO <sub>3</sub>	8.366
MgO	0.117	—	
NaOH	2.124	Na <sub>2</sub> CO <sub>3</sub>	10.053
Ni(OH) <sub>2</sub>	0.498	—	
FePO <sub>4</sub> · <i>x</i> H <sub>2</sub> O	2.711	—	_
PbO	0.401	_	_
Na <sub>2</sub> SO <sub>4</sub>	0.347	—	_
SiO <sub>2</sub>	4.797	SiO <sub>2</sub>	19.643
TiO <sub>2</sub>	0.010	—	_
ZnO	0.081	—	_
Zr(OH) <sub>4</sub> ·xH <sub>2</sub> O	0.989	—	_
H <sub>2</sub> O	105.579		
Na <sub>2</sub> CO <sub>3</sub>	0.304	—	_
NaNO <sub>2</sub>	0.336		
NaNO <sub>3</sub>	0.954		
$H_2C_2O_4{\cdot}2H_2O$	0.115	—	—
2			
Simulant Total	195.217	Additives Total	80.109
		FEED TOTAL	275.326

\* Target weights adjusted for assay information of starting materials

- Empty data field

#### Table 3.16. Composition of Melter Feed to Produce 100 kg of Target Glass HWI-Al-19 (Target Glass Yield = 500 g/L Feed) from the Al-Limited Waste Simulant Using Al(OH)<sub>3</sub> as the Aluminum Source.

Al-Limited W	aste Simulant	Glass-Formi	ng Additives
Starting Materials	Target Weight (kg) <sup>*</sup>	Starting Materials	Target Weight (kg) *
Al(OH) <sub>3</sub>	37.047	—	
$H_3BO_3$	0.341	H <sub>3</sub> BO <sub>3</sub>	34.089
BaCO <sub>3</sub>	0.070	_	_
Bi <sub>2</sub> O <sub>3</sub>	1.156	—	_
CaO	1.099	CaSiO <sub>3</sub> (Wollastonite)	9.798
CdO	0.025	—	_
$Cr_2O_3$	0.532	—	
NaF	1.483	_	
Fe(OH) <sub>3</sub> (13% Slurry)	48.539	—	
KNO <sub>3</sub>	0.308	—	
Li <sub>2</sub> CO <sub>3</sub>	0.432	Li <sub>2</sub> CO <sub>3</sub>	8.625
MgO	0.121	—	
NaOH	2.190	Na <sub>2</sub> CO <sub>3</sub>	10.364
Ni(OH) <sub>2</sub>	0.514	—	
FePO <sub>4</sub> · <i>x</i> H <sub>2</sub> O	2.795	—	
PbO	0.413	—	
Na <sub>2</sub> SO <sub>4</sub>	0.358	—	
SiO <sub>2</sub>	4.945	SiO <sub>2</sub>	17.276
TiO <sub>2</sub>	0.010	—	
ZnO	0.084	—	
Zr(OH) <sub>4</sub> ·xH <sub>2</sub> O	1.020	—	
H <sub>2</sub> O	91.903	—	
Na <sub>2</sub> CO <sub>3</sub>	0.314		
NaNO <sub>2</sub>	0.346	—	
NaNO <sub>3</sub>	0.984	—	
$H_2C_2O_4$ ·2 $H_2O$	0.119	—	_
2	_	—	
Simulant Total	197.148	Additives Total	80.152
		FEED TOTAL	277.300

\* Target weights adjusted for assay information of starting materials

- Empty data field

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	Test	1	2	2	4
		-		_	
e	Feed Start	2/4/08 8:26	2/6/08 12:15	5/5/08 9:32	5/7/08 14:32
Time	Feed End	2/6/08 11:30	2/8/08 23:00	5/7/08 13:30	5/9/08 16:32
	Interval	51.1 hr	51.1 hr       58.75 hr       52.0 hr         34 min       NA       58 min         50.5 hr       58.75 hr       51.0 hr         39 min       50 min       62 min         1200 °C       1150 °C       1200 °C         9.3 lpm       9.2 lpm       9.1 lpm         Hydroxide       Hydroxide       Hydroxide         HLW-E-Al-27       HLW-E-Al-27       HWI-Al-16         45%       45%       43.6%         854 kg       527 kg       879 kg         500 g/l       500 g/l       500 g/l         0.3596 kg/kg       0.3596 kg/kg       0.358 kg/kg         16.9 kg/hr       9.0 kg/hr       16.9 kg/hr         239 kg       251 kg       225 kg         1351 kg/m²/day       717 kg/m²/day       1371 kg/m²/day	50.0	
Water	Feeding for Cold Cap	34 min	NA	58 min	NA
;	Slurry Feeding	50.5 hr	58.75 hr	51.0 hr	50.0 hr
Fee	ding Interruptions	39 min	50 min	62 min	5 min
Targe	t Glass Temperature	1200 °C	1150 °C	1200 °C	1150 °C
Aver	rage Bubbling Rate	9.3 lpm	9.2 lpm	9.1 lpm	9.0 lpm
	Aluminum Source	Hydroxide	Hydroxide	Hydroxide	Hydroxide
	Target Glass	HLW-E-Al-27	HLW-E-Al-27 HLW-E-Al-27		HWI-Al-16
	Waste Loading	45%	45%	43.6%	43.6%
Feed	Used	854 kg	527 kg	879 kg	617 kg
	Target Glass yield	500 g/l	500 g/l	500 g/l	500 g/l
	Target Glass yield	0.3596 kg/kg	0.3596 kg/kg	0.358 kg/kg	0.358 kg/kg
	Average Feed Rate	16.9 kg/hr	9.0 kg/hr	16.9 kg/hr	12.3 kg/hr
I	Poured	239 kg	251 kg	225 kg	281 kg
ducec	Average Rate*	1351 kg/m <sup>2</sup> /day	717 kg/m <sup>2</sup> /day	1371 kg/m²/day	982 kg/m <sup>2</sup> /day
Glass Produced	Steady State Rate*	1200 kg/m²/day	$700 \text{ kg/m}^2/\text{day}$	$1400 \text{ kg/m}^2/\text{day}$	950 kg/m²/day
Gla	Average Power Use	4.1 kW hr/kg glass	5.4 kW hr/kg glass	4.2 kW hr/kg glass	4.4 kW hr/kg glass

\*- Rates calculated from feed data.

Note: Rates do not take into account the time for water feeding and cold cap burn-off.

	Test	5	6	7	8
	Feed Start	6/2/08 12:52	6/4/08 17:30	6/23/08 7:56	6/25/08 12:15
Time	Feed End	6/4/08 15:30	6/7/08 00:30	6/25/08 11:30	6/27/08 23:30
	Interval	50.6 hr	6/2/08 12:52         6/4/08 17:30         6/23/08 7:56           6/4/08 15:30         6/7/08 00:30         6/25/08 11:30           50.6 hr         55.0 hr         51.6 hr           38 min         NA         64 min           50.0 hr         55.0 hr         50.5 hr           13 min         42 min         38 min           1200 °C         1150 °C         1200 °C           8.6 lpm         8.8 lpm         9.0 lpm           Oxide         Oxide         Hydroxide           HWI-Al-16         HWI-Al-16         HWI-Al-19           43.6%         43.6%         45%           806 kg         510 kg         902 kg           500 g/l         500 g/l         500 g/l           0.363 kg/kg         0.363 kg/kg         0.361 kg/kg           16.1 kg/hr         9.3 kg/hr         17.9 kg/hr           214 kg         254 kg         245 kg           291 kg/m²/day         748 kg/m²/day         1434 kg/m²/day           300 kg/m²/day         700 kg/m²/day         1500 kg/m²/day	59.25	
Water I	Feeding for Cold Cap	38 min	NA	64 min	NA
5	Slurry Feeding	50.0 hr	55.0 hr	50.5 hr	59.25 hr
Fee	ding Interruptions	13 min	42 min	38 min	13 min
Targe	t Glass Temperature	1200 °C	1150 °C	1200 °C	1150 °C
Aver	age Bubbling Rate	8.6 lpm	8.8 lpm	9.0 lpm	9.0 lpm
	Aluminum Source	Oxide	Oxide	Hydroxide	Hydroxide
	Target Glass	HWI-Al-16	HWI-Al-16	HWI-Al-19	HWI-Al-19
	Waste Loading	43.6%	43.6%	45%	45%
Feed	Used	806 kg	510 kg	902 kg	693 kg
	Target Glass yield	500 g/l	500 g/l	500 g/l	500 g/l
	Target Glass yield	0.363 kg/kg	0.363 kg/kg	0.361 kg/kg	0.361 kg/kg
	Average Feed Rate	16.1 kg/hr	9.3 kg/hr	17.9 kg/hr	11.7 kg/hr
Ŧ	Poured	214 kg	254 kg	245 kg	239 kg
duceo	Average Rate <sup>*</sup>	1291 kg/m <sup>2</sup> /day	748 kg/m <sup>2</sup> /day	1434 kg/m²/day	938 kg/m <sup>2</sup> /day
Glass Produced	Steady State Rate*	1300 kg/m²/day	700 kg/m <sup>2</sup> /day	1500 kg/m²/day	950 kg/m²/day
Gla	Average Power Use	glass	4.8 kW hr/kg glass	4.1 kW hr/kg glass	4.6 kW hr/kg glass

#### Table 4.1. Summary of Results from DM100 Tests (continued).

\*- Rates calculated from feed data.

Note: Rates do not take into account the time for water feeding and cold cap burn-off.

### Table 4.2. Steady-State Production Rates Achieved with HLW Compositions on the DM100 at Melt Pool Bubbling of 9 lpm and Solids content near 500 g glass/liter.

HLW Waste	Glass Yield (g/L)	Glass Temperature (°C)	Production Rate kg/m <sup>2</sup> /day
Aluminum Limited	500	1200	1200
(HLW-E-Al-27, Al hydroxide)	500	1150	700
Aluminum Limited	500	1200	1400
(HWI-Al-16, Al hydroxide)	500	1150	950
Aluminum Limited	500	1200	1300
(HWI-Al-16, Al oxide)	500	1150	700
Aluminum Limited	500	1200	1500
(HWI-Al-19, Al hydroxide)	500	1150	950
Aluminum Limited	500	1175	550
(HLW-E-Al-27, Al oxide)	500	1150	550
Aleminene en l Ce diene Limited [2]	500	1175	900
Aluminum and Sodium Limited [2]	500	1150	400
	500	1175	1000
Bismuth Limited [2]	500	1150	830
	500	1175	1300
Chromium Limited [2]	500	1150	1150
AZ-101 [14]	530	1150	1300
AZ-102, Nominal Rheology [23]	550	1150	1200
AZ-102, Adjusted Rheology [23]	550	1150	1400
C-106/AY-102, SIPP [15]	470	1150	1180
HLW02-24, High Viscosity [24]	500	1150	900
HLW-ALG-17, Low Viscosity [24]	500	1150	1600
HLW-ALG-16, High Conductivity [24]	500	1150	1200
HLW02-46, Low Conductivity [24]	500	1150	900
C-106/AY-102 [24]	500	1150	1000
C-106/AY-102 + 15% GFCs [24]	500	1150	1000
C-106/AY-102- 15% GFCs [24]	500	1150	1050
C-106/AY-102 replacing borax with boric acid and soda ash [20]	500	1150	1050
C-106/AY-102, SiO <sub>2</sub> = 33% [25]	500	1150	2000
$C-106/AY-102$ , $SiO_2 = 53.1\%$ [25]	500	1150	700
C-106/AY-102, $MnO = 8% [25]$	500	1150	750
$C-106/AY-102$ , $Na_2O = 20.0\%$ [25]	500	1150	2150
C-106/AY-102, $K_2O = 2.5\%$ , $Cr_2O_3 = 0.6\%$ , La <sub>2</sub> O <sub>3</sub> = 1.2%, TiO <sub>2</sub> = 1%, ZnO = 4% [25]	500	1150	1700
$\frac{124203}{\text{C}-106/\text{AY}-102, \text{ B}_2\text{O}_3 = 4.3\% [25]}$	500	1150	900
$\frac{\text{C-106/A1-102, B}_{2}\text{O}_{3} = 4.5\% [25]}{\text{C-106/AY-102, B}_{2}\text{O}_{3} = 15\% [25]}$	500	1150	1550
$\frac{\text{C-106/A1-102, B}_{2}O_{3} = 13\% [25]}{\text{C-106/AY-102, Al}_{2}O_{3} = 13\% [25]}$	500	1150	900
C-106/AY-102, Ai <sub>2</sub> O <sub>3</sub> = 15% [25]	500	1150	450
Algorithm Generated, $ZrO_2 = 10.65\%$ [25]	500	1150	850
Algorithm Generated, $210_2 = 10.03\%$ [25] Algorithm Generated, SrO = 9.27% [25]	500	1150	650
Argonulli Generated, $SIO = 9.27\% [25]$	500	1150	030

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	Test			1		2		3			4			
			AVG	MIN	MAX									
		East Upper	1076	677	1155	1096	1073	1113	988	335	1156	1089	1059	1112
	Electrode	West Upper	1138	773	1186	1095	1073	1127	1088	695	1178	1111	1089	1140
Т	Lieuloue	West Lower	1125	1104	1167	1063	1051	1093	1119	1089	1139	1072	1056	1100
Е		Bottom	804	794	841	759	748	794	726	718	753	705	694	729
M P		27" from bottom	1076	192	1195	1131	1055	1156	1014	118	1207	1139	1099	1160
E	Glass	16" from bottom	1159	773	1216	1141	1115	1167	1126	309	1213	1149	1125	1166
R	Glass	10" from bottom	1203	1179	1242	1153	1132	1178	1201	1162	1223	1155	1135	1176
A		5" from bottom	1192	1137	1229	1147	1126	1169	1191	1155	1219	1140	1122	1163
T U	Plenum	Exposed	468	246	664	449	383	573	437	302	607	406	314	602
R	Tienum	Thermowell	442	356	656	415	370	516	395	272	588	392	313	625
Е	Discharge	Chamber	1022	806	1075	1038	1023	1060	1056	1008	1095	1060	1022	1080
(°C)	Discharge	Air Lift	1023	913	1109	1029	1009	1069	1049	927	1130	1054	1036	1101
	Film C	Cooler Outlet	285	260	301	283	277	287	287	262	306	283	266	296
	Transiti	on Line Outlet	271	219	288	269	265	276	270	218	287	268	211	278
	Lance Bubbling (lpm)		9.3	1.6	9.4	9.2	9.0	9.4	9.1	8.6	9.3	9.0	9.0	9.2
Μ	Melter Pressure (inches water)		-1.05	-3.38	0.44	-1.17	-3.44	0.62	-1.03	-4.34	0.27	-0.99	-1.62	0.50
Т	Total Electrode Voltage (V)		46.0	1.8	57.3	42.5	38.3	50.0	49.2	1.9	56.7	46.3	35.8	48.9
	Total Pov	wer (kW)	24.9	0.3	32.6	17.4	13.9	22.0	25.4	0.3	29.9	19.3	12.8	21.4
	Glass Resist	ance (ohms)	0.085	0.011	0.105	0.104	0.095	0.117	0.096	0.012	0.134	0.111	0.100	0.121

#### Table 4.3. Summary of Measured DM100 Parameters.

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	Test			5			6		7			8		
			AVG	MIN	MAX									
		East Upper	1044	719	1151	1072	1010	1105	1042	482	1156	1087	1067	1141
	Electrode	West Upper	1083	758	1153	1073	1019	1118	1107	716	1170	1103	1084	1169
Т	Electione	West Lower	1126	1108	1144	1062	1035	1093	1134	1114	1150	1081	1068	1143
Е		Bottom	716	650	734	699	689	729	714	655	733	702	693	733
M P		27" from bottom	1129	721	1209	1138	1094	1160	1058	323	1202	1120	1075	1176
E	Glass	16" from bottom	1187	867	1221	1149	1120	1170	1118	240	1204	1133	1096	1183
R	01455	10" from bottom	1202	1177	1227	1157	1123	1185	1190	1150	1214	1146	1122	1200
A		5" from bottom	1195	1163	1217	1135	1103	1168	1202	1173	1219	1153	1132	1210
T U	Plenum	Exposed	462	316	649	412	157	621	505	417	655	446	345	626
R	1 Ichum	Thermowell	417	286	631	378	156	663	468	397	622	413	343	596
Е	Discharge	Chamber	1063	1039	1097	1058	1022	1081	1061	1004	1099	1060	978	1084
(°C)	Discharge	Air Lift	1061	983	1134	1048	1024	1120	1065	962	1162	1062	1007	1115
	Film <b>(</b>	Cooler Outlet	286	256	307	275	241	292	285	281	296	279	268	287
	Transiti	on Line Outlet	272	219	291	263	229	275	273	232	283	268	213	274
	Lance Bub	bling (lpm)	8.6	1.1	8.9	8.8	8.6	9.0	9.0	1.5	9.1	9.0	8.9	9.5
Μ	Melter Pressure (inches water)		-0.90	-2.29	0.31	-0.99	-3.66	0.31	-0.97	-3.20	0.80	-0.99	-2.86	0.38
Т	Total Electrode Voltage (V)		48.2	1.2	57.0	43.0	39.1	49.0	47.2	1.2	57.0	41.5	28.9	45.2
	Total Power (kW)		23.9	11.6	28.0	16.3	13.9	20.0	26.3	10.6	28.2	18.8	10.0	26.0
	Glass Resist	ance (ohms)	0.098	0.082	0.132	0.114	0.093	0.151	0.086	0.072	0.136	0.092	0.078	0.101

#### Table 4.3. Summary of Measured DM100 Parameters (continued).

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	-	1	2	3
	Feed Start	8/6/08 9:21	8/11/08 14:45	8/13/08 16:05
Time	Feed End	8/8/08 10:21	8/13/04 16:00	8/15/08 19:30
	Interval	49.0 hr	49.25 hr	51.4 hr
Water F	eeding for Cold Cap	1.0 hr	1.0 hr	NA
S	lurry Feeding	48.0 hr	48.25 hr	51.4 hr
(	Cold cap burn	3.2 hr	NA	4.0 hr
Avera	ige Total Bubbling	125 lpm	65 lpm	54 lpm
Stead	dy State Bubbling	124 lpm	71 lpm	48 lpm
	Average Glass mperature, East	1150°C	1154°C	1178°C
	Average Glass nperature, West	1148°C	1146°C	1170°C
	verage Plenum Temperature	653°C	571°C	595°C
Avera	ge Electrode Power	225 kW	170 kW	181 kW
E. J	Used	9942 kg	6932 kg	7498 kg
Feed	Average Rate	207.1 kg/hr	143.7 kg/hr	145.9 kg/hr
pa	Poured	3737 kg	2404 kg	2732 kg
roduc	Average Rate <sup>\$</sup>	1557 kg/m²/day	996 kg/m²/day	1063 kg/m <sup>2</sup> /day
Glass Produced	Average Rate*	1495 kg/m²/day	1038 kg/m²/day	1053 kg/m²/day
Gl	Steady State Rate*	1500 kg/m²/day	1050 kg/m²/day	1050 kg/m²/day

#### Table 5.1. Summary of DM1200 Test Conditions and Results.

\$ - Rates calculated from glass poured.

\*- Rates calculated from feed data.

Note: Rates do not take into account the time for water feeding and cold cap burn-off. NA: Not applicable.

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	Operational notes					
Test	Run time (hours)	Run time note				
	0.0	Start water feeding at 8/6/2008 9:21				
	1.0	Started slurry feeding				
	5.3	Paused feeding for 5 minutes to collect feed sample.				
	7.5	During the SBS blowdown, Y-strainer clogged and was cleaned out.				
	10.5	Transferred feed from mixing tank to feed tank.				
	10.9	Film cooler rinse was performed.				
	11.0	During film cooler rinse, it was discerned the film cooler was ~90% clogged. Paused feeding for 8 minutes to manually clear film cooler.				
	20.2	Transferred feed from mixing tank to feed tank.				
	23.4	Film cooler rinse was performed.				
	26.7	Took picture of cold cap through melter view port. (see Figure 5.1)				
	27.4 - 27.5	Performed WESP deluge, pre-blowdown and post blow down.				
1	30.4	Transferred feed from mixing tank to feed tank.				
1	33.4	Paused feeding for 4 minutes to collect feed sample.				
	35.0	Melter tripped into Emergency Off Gas line because port was opened to remove exhaust sampling probe. Exhaust immediately returned to main line.				
	35.6	Film cooler rinse was performed. Film cooler blockage was ~40-50%. Film cooler rinse opened up the blockage. Below the film cooler there was still residual blockage of about 10% after rinsing.				
	40.7	Transferred feed from mixing tank to feed tank.				
	49.00	Feed stopped at the end of Test 1.				
	49.6	Opened port C-1 (see Figure 1.6) to retrieve a cold cap sample. The gasket ring was replaced.				
	49.9	Film cooler rinse was performed.				
	51.0	Feed transferred to mixing tank for Test 2.				
	51.9 - 60.0	Performed WESP deluge, pre-blow down and post blow down.				
	52.2	Off gas shut down is completed.				

#### Table 5.2. Summary of Operational Events.

	Operational notes					
Test	Run time (hours)	Run time note				
	-	Transferred feed from mixing tank to feed tank.				
	0.0	Start water feeding at 8/11/2008 14:45				
	1.0	Slurry feed started				
	7.4	Paused feeding for 6 minutes to collect feed sample.				
	10.6	Film cooler rinse was performed.				
	14.9	Transferred feed from mixing tank to feed tank.				
	18.1	Transferred feed to mixing tank.				
	22.3	Paused feeding for 6 minutes to collect feed sample.				
	23.1 - 23.3	Blower 701 failed. Paused feeding for 13 minutes to switch to Blower 702.				
	23.5	Melter tripped into Emergency Off Gas line because port was opened for removal of exhaust sampling probe. Exhaust immediately returned to main line.				
2	23.6 - 23.9	Performed WESP deluge, pre-blow down and post blow down.				
	23.8	Film cooler rinse was performed.				
	25.3	Melter tripped into Emergency Off Gas line due to port opened to remove exhaust sampling probe. Exhaust immediately returned to main line.				
	29.6	Transferred feed from mixing tank to feed tank.				
	34.9	Film cooler rinse was performed.				
	41.2	Replaced Blower-701 head for back-up. Observed that Blower-702 outlet temperature is 90°C. This could cause premature failure. Reduced Blower setting from 46 to 45 Hz and increased Blower 801setting from 21 to 23 Hz.				
	44.3	Transferred feed from mixing tank to feed tank.				
	47.3	Film cooler rinse was performed.				
	48.1 - 48.2	Performing WESP deluge, pre-blow down and post blow down.				
	49.2	Film cooler rinse was performed.				
	49.3	Feed stopped at end of Test 2. Feed sample was also collected.				

#### Table 5.2. Summary of Operational Events (continued).

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	Operational notes					
Test	Run time (hours)	Run time note				
	49.3	Changed test conditions at 8/13/2008 16.05				
	56.5 - 56.6	Film cooler differential pressure was up, 9.5"WC, film cooler was cleared.				
	58.3	Transferred feed from mixing tank to feed tank.				
	63.9 - 64.1	Feed tube was partially clogged. Tapped with hammer to dislodge the blockage. Feed was paused for 13 minutes. Feed sample was collected.				
	64.6	Electric high temperature alarm sounded (East Electrode thermocouple reading, TR-21 =1172 °C). Reduced power from 150 to 145 kW, later electric was shut down. Increased lance bubbler 2 flow rate from 20 to 22 lpm.				
	64.9	Electric alarm was cleared. Sounded. East Electrode thermocouple reading was 1164 °C.				
	66.0	Transferred feed from mixing tank to feed tank.				
	66.2	Cold cap ridge or cone collapsed, causing pressure surge. Emergency Off Gas was activated. Exhaust immediately returned to main line.				
	71.6 - 71.8	Performed WESP deluge, pre-blow down and post blow down.				
	73.1	Transferred feed from mixing tank to feed tank.				
3	73.2	Film cooler rinse was performed.				
5	77.10	High Temperature alarm and power shut off set points attached to the East Electrode thermocouple were increased to 1175 °C and 1180 °C respectively.				
	84.8	Film cooler rinse was performed.				
	86.3	Transferred feed from mixing tank to feed tank.				
	88.4	Feed tube was manually flushed.				
	92.5	Paused feeding for 7 minutes to collect feed sample.				
	96.7	Film cooler rinse was performed.				
	99.7 and 100.7	Tapped on the feed tube to clear the stalactite which was growing under the feed tube.				
	100.8	Feed stopped at the end of Test 3.				
	102.0	Removed feed from feed tank.				
	106.0 - 106.4	Performed WESP deluge, pre-blow down and post blow down.				
	107.6	Started melter and off-gas shut down.				

Table 5.2. Summary of Operational	l Events (continued).
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Test	Date	Time	Run Time (hours)	Cold Cap Observations
		9:21	0.0	Started feeding water at 500 mL/min.
		11:03	1.7	CC ~80% is flat with boiling on top and feed flowing into opening in S-E&N-W corners.
		11:15	1.9	CC 75-80% with liquid boiling profusely and flowing into openings in between feed shots.
		11:34	2.2	CC ~75-80% with liquid boiling profusely.
		11:50	2.5	CC opening on east side of melter is larger than west.
		12:05	2.7	CC ~80% with liquid boiling.
		12:11	2.8	CC is ~85% with liquid flowing to openings in between feed shots.
		12:33	3.2	CC $\sim$ 85%. Flat $\sim$ 6" thick with feed boiling on surface. Opening on S-E and N-W corners with feed flowing into the openings.
		13:05	3.7	CC ~85% ~6" thick with small ridge around openings on S-E & N-W corners. Feed boiling on the surface and flowing into openings.
		13:12	3.9	CC unchanged.
		13:40	4.3	Increase bubbling from 60 to 80 lpm, CC ~85% with more aggressive liquid boiling observed.
		13:53	4.5	CC is still ~80-85 % with liquid boiling and flowing into openings.
		14:07	4.8	CC~85 % with liquid mildly boiling and flowing into openings.
		14:19	5.0	CC $\sim$ 80% with opening slightly larger and center island bridge $\sim$ 8" thick liquid on each side.
1	8/6/2008	14:43	5.4	CC~80-85 and 6-8" thick boiling liquid pool and center ridge.
		15:05	5.7	CC ~80-% couple of high ridges around openings.
		15:35	6.2	CC ~80% and 6-8" thick.
		15:56	6.6	CC ~90%. Can not see east opening due to center ridge.
		16:13	6.9	CC ~90%, No visible changes other than larger amount of liquid is flowing into melt pool.
		16:35	7.2	CC ~85%. Feed is slow to boil but is boiling in between each feed shot.
		16:49	7.5	$CC \sim 85$ %, Not able to see east side anymore due to ridge on that side.
		17:10	7.8	CC $\sim$ 80%, large ridge appears to be near center of melter, can not see east side.
		17:26	8.1	CC ~75%, large ridge is still present as well as build up on the west wall.
		17:38	8.3	CC $\sim$ 75%, ridge seems smaller and vigorously boiling feed can be seen on the east side of the melter.
		17:53	8.5	CC ~75%, appears to be the same as last observation.
		18:08	8.8	CC $\sim$ 75%, appears as before but build up on the west wall which is just south of middle view port extends out past but does not yet touch T/W 2.
		18:22	9.0	CC ~75%.
		18:37	9.3	CC ~75%, build up has now surrounded T/W #2, otherwise conditions appear unchanged.
		18:40	9.3	Increased bubbling on Lance 1 from 55 to 85 lpm, try to shift the cap around to break the ridges.

 Table 5.3. Operator Observations of Cold Cap (CC).

Test	Date	Time	Run Time (hours)	Cold Cap Observations
		18:50	9.5	Increased Lance 2 from 65 to 85 lpm and at Lance 1 reduced bubbling from 85 to 55 lpm. Trying to shift cap/ridges around.
		18:55	9.6	CC ~80%. The opening on west side has closed up some. Feed is boiling over the ridge. Not able to see east side at this time.
		19:09	9.8	CC $\sim$ 80%, Since discharge cap has not moved ridge looks bigger, still not able to see east side other than the light that is emitting from that side. Reduced Lance 2 from 85 to 55 lpm, and increased Lance 1 from 55 to 65 lpm.
		20:12	10.9	CC ~80%, large ridge on the west side is high not allowing to see east side. Increased bubbling on Lance 2 from 65 to 90 lpm to see if this will break it down.
	8/6/2008	20:41	11.3	CC ~75%, Glass level is low. Increased bubbling on Lance 1 to 85 from 65 lpm and reduced bubbling on Lance 2 from 90 to 65 lpm.
		21:08	11.8	CC ~75% very little or no change.
		22:13	12.9	CC ~85%, ridge still divides east and west. East side opening can not be seen. Nothing can be seen through middle view port.
		22:28	13.1	CC ~85% as before.
		22:55	13.6	CC ~80-85%, opening visible around Lance 2 with slight ridge around. Light is visible on east wall.
		23:10	13.8	CC~80-85%, basically unchanged.
		23:32	14.2	CC ~75 - 80%.
1		23:40	14.3	CC ~80%. Ridge on the west side seems lower.
1		0:08	14.8	CC is at 85%, ridge is much smaller around Lance #2, light is still visible on east.
		0:53	15.5	CC ~85 %, large ridge in center of melter.
		1:42	16.4	CC ~85 %, ridge has grown larger near Lance #2.
		2:23	17.0	CC ~85 %, but ridge in center/ seems to be growing with increased bubbling.
		2:37	17.3	CC ~85%, only opening on the west wall partially visible.
		2:59	17.6	From the south view port, "existing wall like" portion of cold cap, positioned just below the feed tube and stands $<$ or=12".
	8/7/2008	3:39	18.3	CC ~85%, and no visible change.
		3:58	18.6	Larger mound growing in melter seen via south view port, also large glazed ridge near Lance #2.
		4:25	19.1	Raised and lowered bubbling on both Lance's in an effort to break down large mound in south view port with no success.
		4:51	19.5	Large mound is still in place around Lance #2.
		5:04	19.7	Mound is still in place.
		5:17	19.9	CC ~75-80%, mound is still in place but starting to see more liquid feed trickling off of ridge by Lance #2 and flowing into melt pool.
		5:34	20.2	CC ~80%, mound still in place, however, ridge around Lance #2 has broken down some.
		5:49	20.5	CC ~80-85%.

Table 5.3. Operator Observations of Cold	l Cap (CC) (continued).
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Test	Date	Time	Run Time (hours)	Cold Cap Observations	
		6:02	20.7	CC ~80-85%.	
		7:05	21.7	CC ~85%, Large ridge in center of melter. Openings on S-E and N-W corners. Large mound has formed around feed tube.	
		7:14	21.9	CC ~95%, large ridge and funnel below feed tube. Openings on S-E and N-W corners.	
		7:25	22.1	CC $\sim$ 90-95%, with large ridge in center with funnel below feed tube. Openings at S-E and N-W corners.	
		7:50	22.5	CC ~90%, with high center. Light visible on east side. Liquid visibly flowing into L2 bubbler. Opening on west side.	
		8:10	22.8	CC conditions are stable ~90-95 % with liquid flowing.	
		8:22	23.0	CC ~95%, liquid flowing into openings. Light visible on east side.	
		8:37	23.3	No visible change in CC conditions.	
		9:15	23.9	CC ~95% with large ridge in center with openings on S-E and N-W corners.	
		9:40	24.3	CC ~90-95% with large ridge in center with openings in S-E and N-W corners. Feed visibly flowing into N-W opening. Funnel below feed tube has not changed.	
		10:10	24.8	CC ~90-95%, large ridge in center with openings in S-E and N-W corners. Feed is flowing into N-W corner.	
		10:45	25.4	CC is at 90-95 % with same conditions as before.	
		11:10	25.8	CC conditions unchanged (90-95%)	
1	8/7/2008	11:50	26.5	CC ~90-95% with large ridge in center with openings on S-E and N-W corners. Feed flowing into N-W corner opening.	
		12:00	26.7	Collected picture of the cold cap thru. N. viewport showing part of cap thickness.	
		12:20	27.0	CC ~90%, Large ridge in center with openings in S-E and N-W corners feed is flowing into N-W corner.	
		12:40	27.3	Unchanged from previous observation (CC ~90-95).	
		13:14	27.9	CC ~95%, Opening at Lance 2 has a "glazed" face. Cone below feed tube has grown, it is now almost out of view at top	
		13:20	28.0	Dislodging cone below feed tube manually. Minor pressure spike.	
		13:29	28.1	CC ~95%.	
		13:44	28.4	CC ~95%, can't see liquid flow to Lance 2 opening. Observed splash from boiling liquid through south viewport prior to feed shot.	
		14:00	28.7	CC ~95%.Some liquid now flowing into Lance 2 opening.	
		14:17	28.9	CC ~86%, liquid boiling in center and flowing to Lance 2 opening.	
		14:34	29.2	CC $\sim$ 90%, large mound seen in the south view port. North view port able to see the melt pool with a large ridge blocking the east side.	
		14:50	29.5	CC ~90%, No visible changes at this time.	
		15:02	29.7	Increased Lance 1 from 60 to 80 lpm, trying to collapse the large mound underneath feed tube.	
			15:08	29.8	CC ~90%, starting to see more glow from east side. No change on west side.

Test	Date	Time	Run Time (hours)	Cold Cap Observations
		15:20	30.0	CC ~90%. The cap dropped when glass was discharged. The mound looks smaller due to this.
		15:39	30.3	CC ~90%, west side cap/mound is starting to melt down.
		15:54	30.6	CC ~85%, mound on west side continues to melt.
		16:09	30.8	CC ~90%, more liquid is flowing into the west opening.
		16:16	30.9	Not much has melted down from the mound on the west side. CC~90%.
		16:39	31.3	CC $\sim$ 90%. Not much light seen emitting from east side through the south view port. West side mound has built up some over the opening.
		16:48	31.5	CC ~90%.
		17:13	31.9	CC ~90%.
		17:26	32.1	CC ~90%, able to see feed boiling between shots. Mounds and shots have not grown any.
		17:42	32.4	CC $\sim$ 90%, some melting of mound on west side. Very little changes elsewhere.
		17:55	32.6	CC $\sim$ 90%, The face of the west side mound looks glazed now. East side unchanged.
		18:11	32.8	CC ~90% same as before.
		18:26	33.1	CC ~90%, large glazed ridge can be seen in north port, moderate size circular ridge visible in south port.
1	8/7/2008	18:52	33.5	CC ~90%, no visible changes at this time.
1	8/7/2008	19:10	33.8	CC ~90%. There is large cone shaped mound underneath the feed tube. It has gotten larger and taller.
		19:17	33.9	Melter pressure spiked almost positive due to off-gas sampling.
		19:37	34.3	CC ~90%, conditions have not changed.
		19:54	34.6	$CC \sim 90\%$ , North side mound is melting. It does not look as glazed from previous observation.
		20:11	34.8	CC ~95%, Not much light coming from east side.
		20:20	35.0	CC ~ 90%.
		20:38	35.3	CC ~90, Able to see more light on the east side. West side mound is glazed over melt pool is smaller with a secondary cap forming.
		20:55	35.6	CC $\sim$ 90%, a visible dome of hardened feed in west end of melter. Melt pool is visible from east not much cap.
		21:29	36.1	CC ~85%, a very large glazed wall of glass is visible through north view port. Cylinder like mound which is large can be seen from south port.
		21:43	36.4	CC ~85%, conditions are same as previously noted.
		22:06	36.8	CC ~85%, appears unchanged.
		22:56	37.6	CC ~80-85%, mound is still in place when viewed from south port and glazed ridge near Lance #2 although feed is now flowing into opening.
		23:13	37.9	CC ~80%, large opening still present near Lance #2. Mound unchanged.
		23:39	38.3	CC unchanged since previous observation.

Table 5.3. Operator Observatio	ns of Cold Cap (CC) (continued).
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Test	Date	Time	Run Time (hours)	Cold Cap Observations
	8/7/2008	23:58	38.6	CC ~80-85 % unchanged.
		0:33	39.2	CC ~80% unchanged.
		1:09	39.8	CC ~80% unchanged.
		1:27	40.1	CC $\sim$ 80-85 % seem like mound has grown taller while ridge has broken down somewhat.
		2:12	40.9	Mound still growing, CC at 85%.
		2:55	41.6	CC ~80-85%, unchanged.
		3:17	41.9	CC ~80%.
		3:25	42.1	Increasing Lance from 80 to 120 lpm did not have any effect on mound. Mound is growing close to melter ceiling.
		3:39	42.3	Mound is large near ceiling
		4:08	42.8	Manually dislodged top of mound allowing feed to flow to melt pool. A byproduct of this seems that glass temps have come back.
		4:23	43.0	Attempted once again to dislodge some of the mound but unable to make contact with it. $CC \sim 85\%$ ridges has broken down some.
		5:43	44.4	CC ~85%, ridge near Lance 2, has some feed flowing over it into melt pool. Majority of mound still evident via south view port.
1		6:27	45.1	No light visible.
	8/8/2008	6:35	45.2	$CC \sim 85\%$ , ridge near Lance 2 shows some glazing but liquid flowing into opening. Part of mound still evident via south viewport.
		6:55	45.6	CC ~85%, some liquid flowing to L2 opening. A small amount of light is visible from east side.
		7:10	45.8	Cold cap unchanged (95%).
		7:25	46.1	CC $\sim$ 95% with liquid flowing to L2 opening. East side shows some light reflection on wall.
		7:40	46.3	Cold cap unchanged.
		7:51	46.5	Observing liquid in center "cone" area is boiling more vigorously prior to shot the shot with bubbling increased.
		8:06	46.8	Opening around L2 shows a more glazed surface to cap material with no liquid flow. Center area still boils vigorously, covered still ~95%.
		8:27	47.1	CC $\sim$ 95%, L2 bubbling appear to be undercutting the cap, leaving an overhang. No liquid appears to be flowing to opening. Center cone liquid still reaching vigorous boil prior to shot. Some light visible on east side.
		8:43	47.4	CC ~90-95%, unchanged visually. 10-12 "thick with boiling liquid in center.
		9:52	48.5	Some light on east side.
		10:06	48.8	CC unchanged. 90-95%, 10-12" thick.
		10:21	49.0	End of Test 1.

Table 5.3. Operator	r Observations of Col	ld Cap (CC) (continued).
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Test	Date	Time	Run Time (hours)	Cold Cap Observations
		10:55	49.6	Removed port C-1 to retrieve cold cap sample, Very little material on surface. Scraped collector along hard shell material for sample. O-12J-106A, mass=268 g. Port C-1 gasket needs a new gasket ring and ring was replaced.
		11:05	49.7	Shelf/CC still ~80%, and 10-12" thick.
		11:18	50.0	Shelf beginning to dissolve and expose cavities within. Still thick at center with cone still in place $\sim 12''$ at thickest point.
		11:26	50.1	Shelf has dropped from bottom side and provided a partial (~30%) cold cap with 2-3" separation to a shelf/bridge.
		11:45	50.4	Shelf dissolving from bottom but still bridged over melt pool.
		12:05	50.7	Large clumps of cold cap/bridge pieces in S-W and N-E corners. Openings in middle and S-E and N-W corners. Observed 2 mounds on south side of pool ~8-10" thick, no mound or bridge observed through N. viewport. About 5 % feed material dissolving on glass surface.
		12:25	51.1	Mound in south area diminished. But still visible through S. viewport. Suggesting 6-8" thick.
		12:37	38.7	Mounds of material have dropped to melt surface. Mounds are $\sim$ 1-12" in diameter and $\sim$ 4-6" thick.
1	8/8/2008	11:05	49.7	Shelf/CC still ~80%, and 10-12" thick.
		11:18	50.0	Shelf beginning to dissolve and expose cavities within. Still thick at center with cone still in place $\sim 12''$ at thickest point.
		11:26	50.1	Shelf has dropped from bottom side and provided a partial (~30%) cold cap with 2-3" separation to a shelf/bridge.
		11:45	50.4	Shelf dissolving from bottom but still bridged over melt pool.
		12:05	50.7	Large clumps of cold cap/bridge pieces in S-W and N-E corners. Openings in middle and S-E and N-W corners. Observed 2 mounds on south side of pool ~8-10" thick, no mound or bridge observed through N. viewport. About 5 % feed material dissolving on glass surface.
		12:25	51.1	Mound in south area diminished. But still visible through S. viewport. Suggesting 6-8" thick.
		12:37	38.7	Mounds of material have dropped to melt surface. Mounds are $\sim$ 1-12" in diameter and $\sim$ 4-6" thick.
		12:58	51.6	Cold cap log of feed material floating on surface.
		13:04	38.7	Cold cap material dissolved down to 3"x10" logs. Two chunks of feed material still floating on surface. One is about the size of a football the other one is about the size of a softball.
		13:17	52.0	Only a small area of darker material on melt surface remains. No feed material solids visible. Cold cap is gone.
		14:45	0.0	Started feeding water for test 2.
		15:55	1.2	CC ~80%. Cold cap is very thin with liquid boiling on surface.
2	8/11/2008	16:10	1.4	CC ~75%, 2 openings on both sides. East side opening is larger than west side's. Cold cap is flat at this time.
		16:23	1.6	CC ~80, Starting to get thicker.
		16:37	1.9	CC ~80%, it is a little thicker than last observation.

Table 5.3. Operator	<b>Observations of Cold</b>	Cap (CC) (continued).
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Test	Date	Time	Run Time (hours)	Cold Cap Observations	
		16:45	2.0	CC ~80%, cap hot along the walls. Heavy boiling on the center observed.	
		17:01	2.3	CC ~75%, it is starting to stiffen up. Larger amount of feed boiling on the surface.	
		17:15	2.5	CC ~80%, Starting to close up some. There are small ridges around the openings but feed is flowing into the melt pools.	
		17:30	2.8	CC $\sim$ 80%. The ridges are getting larger. The openings are staying the same size.	
		17:45	3.0	CC ~80%, Hot cap around Thermowell #2. No other changes at this time.	
		18:01	3.3	CC ~85%. Cap has closed up some on east side. On west side no visible changes.	
		18:17	3.5	CC ~90-95% with small ridges around openings. Cap is 2-3" thick with liquid boiling vigorously.	
		18:32	3.8	There is more liquid rushing into the melt pools at this time. The ridges are the same as last entry.	
		18:39	3.9	Cold cap has opened up some on the west side.	
		18:53	4.1	Cold cap ~90%, Ridge around the openings have grown some 3-4" high. Moderate amount of liquid boiling all the time.	
		19:10	4.4	CC ~85%. Cap is floating on the melt pool.	
		19:27	4.7	CC $\sim$ 90%, Not able to see the east side.	
2	8/11/2008	19:42	5.0	CC ~90%. More liquid flowing into the west side melt pool. Only see light emitting from the east side through the south view port.	
		20:05	5.3	CC ~90%. Able to see a small opening on the east side. West side cap around Thermowell #2 has darkened up.	
		20:22	5.6	CC ~95%. West side has closed up more.	
		20:36	5.9	CC ~90%. Not able to see east side. West side has a ridge that curls back to the center. Feed is boiling over it	
		20:53	6.1	CC ~90%. Conditions unchanged.	
			21:06	6.4	CC ~90%, Conditions are unchanged except that it looks like more feed is pooling on the west side.
		21:24	6.7	CC ~90%, Ridge is larger now on the west side but there is no ridge close to the wall. Feed is boiling over. Not able to see east side.	
		21:39	6.9	CC ~90, There is more liquid feed flowing into the west melt pool.	
		21:47	7.0	CC ~90, west side is starting to melt down due to the higher bubbling.	
		22:17	7.5	CC ~85%, West side opened up some.	
		22:32	7.8	CC ~90%, West side closed up some.	
		22:46	8.0	CC at 85-90 % fairly flat on west side, light is visible on east side.	
		23:07	8.4	CC ~85%, with liquid spilling into opening on west side.	
		23:34	8.8	CC~85-90% unchanged.	
		23:56	9.2	CC ~85% opening on west side appears larger.	

Table 5.3. Operator	· Observations of Co	ld Cap (CC) (continued).
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Test	Date	Time	Run Time (hours)	Cold Cap Observations
		0:10	9.4	CC ~80-85% opening on west side is starting to glaze up, not much liquid trickling into opening.
		0:25	9.7	CC ~80-85%, more feed starting to flow into west opening. Light is visible on east side.
		0:40	9.9	CC ~85%, West side opening still has a thick ridge around it. But glazing is reduced. Light still on east side.
		0:55	10.2	CC ~90%, opening on west side closed considerably.
		1:33	10.8	CC ~85%, Ridge appears larger on west side, small amount of feed rushing into opening.
		1:51	11.1	CC ~90%, Opening on west side is almost closed, but also has about 5" or - 6" ridge around it. Light still on east.
		2:10	11.4	CC ~90%, unchanged since previous observation.
		2:30	11.8	CC ~85% same as previous observation.
		2:50	12.1	CC ~85% unchanged since previous.
		3:05	12.3	CC ~85%, same as previous.
		3:48	13.1	Not seeing any feed flowing over ridge currently has a large opening along west wall. CC $\sim$ 80%.
		4:06	13.4	CC 90-95%, finally big rush of feed along west wall completely dark on that side.
		4:32	13.8	CC ~90-95%, slight amount of light visible.
2	8/12/2008	4:58	14.2	CC ~90%, visible light evidence of opening on east, partial opening on the west. Portion of the cold cap surface shows some liquid and boiling.
		5:48	15.1	CC ~80 - 85%.
		6:06	15.4	North Port: Cold cap portion with liquid boiling about 10" towards the center is a thicker portion limiting further view. South Port: Light from the east wall, no mounds at the center could be seen.
		6:26	15.7	CC ~90%, opening visible on west side with lower ridge in center. Light is visible on east side.
		6:40	15.9	CC ~90%, large mound in center, blocking view to east side through North view port.
		7:11	16.4	CC unchanged in appearance, still 90-95%.
		7:23	16.6	CC ~90%, liquid observed flowing into Lance 2 opening and splashing from boiling in center area via south view port.
		7:45	17.0	Cold cap thickness appears to be $\sim 12 - 14"$ in center are, currently it appears most of the feed is flowing to Lance 2 opening.
		8:14	17.5	No change in cold cap appearance, still 90-95% coverage and 10-14" thick at the center.
		8:33	17.8	CC unchanged.
		9:13	18.5	CC unchanged except the surface of opening in Lance 2 area has a more "glazed" glass coating.
		9:37	18.9	CC appearance unchanged at 90-95%, 10-14" thick.
		10:12	19.5	$CC \sim 90-95\%$ with feed flowing predominantly into Lance 2 opening

Table 5.3. Operator Observations of Cold	Cap (CC) (continued).
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Test	Date	Time	Run Time (hours)	Cold Cap Observations
		10:38	19.9	CC ~90-95%.
		11:13	20.5	CC 90-95%, 10-14" thick.
		12:10	21.4	CC conditions are unchanged.
		12:15	21.5	CC has softened around thermowell #2.
		12:37	21.9	CC ~90-95 %, openings are on S-E and N-W corners. Ridge ~14" thick as seen from N-W view port. Light is visible in S-E corner via S-W view port.
		13:21	22.6	CC ~90%.
		13:36	22.9	CC ~80-85%.
		13:44	23.0	CC ~80 with no feed flowing to Lance 2 area.
		14:23	23.6	CC ~85%. Feed is starting to flow into the west side of the melt pool. Only see light emitting from the east side.
		14:39	23.9	CC ~90%, very little light on east side, slightly more on west side. Nothing can be seen in center view port. Large glazed wall has surface that looks crumbly.
		14:48	24.1	CC ~95%. East side has some light emitting from the south view port.
		15:04	24.3	CC ~95% still power limited, set point is 155 kW, output is 141 kW. West side has large amount of feed flowing toward the west wall.
2	8/12/2008	15:17	24.5	CC ~95%, there are 2 pools of feed that can be seen. The lower pool along the west wall. A ridge is holding it back until it boils over. Not able to see east side.
_	0,12,2000	15:38	24.9	CC ~95%.
		15:52	25.1	CC ~95%, no visible changes since last observation.
		16:17	25.5	CC ~95%, not able to see east side. West side glass is being splashed up.
		16:35	25.8	CC ~95%.
		17:10	26.4	CC ~95%, there is a hole in the side of the ridge allowing feed to flow through into the melt pool.
		17:25	26.7	CC ~95%, Able to see a cone mound rising underneath the feed tube. No other visible changes.
		17:39	26.9	CC ~95%, No changes from last observation.
		17:48	27.1	CC ~90%, cap opened up a little on west side. Not much light is coming from east.
		17:53	27.1	CC opened up on west side.
		18:16	27.5	CC ~85%, west side opened up some more. There is a hole in the side of the ridge. Feed is flowing into the opening.
		18:34	27.8	CC ~85%, the several holes in the ridge visible through north view port are getting larger., feed is boiling on the surface and flowing into the glass pool through the south view port of top part of an inverted cone can be seen.
		18:50	28.1	CC ~85%, holes in the ridge are getting larger, some areas of the cap near the middle view part are floating
		19:15	28.5	CC ~85%, The west side ridge is melting.

Table 5.3. Operator	r Observations of Col	ld Cap (CC) (continued).
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Test	Date	Time	Run Time (hours)	Cold Cap Observations
		19:35	28.8	CC ~85%, Starting to close up. There is large amount of liquid feed on cold capthat boils over into the melt pool right before each shot.
		19:51	29.1	CC ~90%, the opening on west side is closing. Around thermowell #2 cold cap stiffening up.
		20:08	29.4	CC ~90%, no visible changes.
		20:43	30.0	CC ~85%, ridge on the west side has stopped melting. It looks dry. Cap is hot around thermowell #2
		21:10	30.4	CC ~90%, more liquid feed is flowing into west melt pool.
		21:24	30.7	$CC \sim 90\%$ , east side able to see light emitting from the south view port. The cone shape mound underneath feed tube has collapsed just around the top. There is a ridge forming over the west side melt pool. Feed is still flowing over it.
	8/12/2008	21:41	30.9	CC ~90%, the lower part of the ridge is visible from north port is thin and has lots of holes and looks like there is a thicker section above it. A mound is growing in front of middle view port. Cone and east side look mostly the same.
		21:57	31.2	CC ~90%, appears unchanged.
		22:42	32.0	CC ~85-90 % with feed flowing into opening on west side of melter mound is visible via south view port.
		22:58	32.2	$CC \sim 80-85$ %, there is large opening on the west side, mound growing in the south view port.
2		23:14	32.5	CC ~80-85%.
		23:34	32.8	CC ~80-85% and ~10 inches high ridge on Lance 2 opening has a hole revealing heavy liquid accumulation on the cap surface. Feed flows to the melt pool between shots. Lance 2 opening is also visible through the middle port. From the south port observable mound is present. Light along the mound L1 exists.
		0:02	33.3	CC ~85%, observed heavy liquid flow from the surface through the hole around Lance 2 opening.
		0:23	33.6	CC ~85%, partial Lance 2 opening visible from the mid-view port is slightly reduced. Others are the same.
		0:40	33.9	CC ~90%, still large amount of feed flowing through hole in center ridge to west side of melter. Large mound still growing.
		1:00	34.3	CC ~90% is same as before.
	8/13/2008	1:12	34.5	CC ~90%, opening on west wall is closing up.
		1:26	34.7	CC at 85%, opening on west side appears larger.
		1:55	35.2	CC ~85% unchanged.
		2:15	35.5	CC ~85-90, west opening unchanged, mound growing taller to south with a stalactite. Growing near feed tube.
		2:22	35.6	Removed top section of mound and stalactite by rodding.
		2:45	36.0	CC ~85%, liquid coming from center of cap formed through-like portion of cap where it flows freely to melt pool

Table 5.3. Operator	· Observations of	f Cold Cap	(CC) (continued).
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Test	Date	Time	Run Time (hours)	Cold Cap Observations
		3:36	36.9	CC ~85-90 %, opening near west wall appears to have a second small ridge growing near it. Mound is still in place.
		4:05	37.3	CC ~85%.
		4:26	37.7	CC ~85-90%, ridge seems slightly larger.
		4:47	38.0	CC ~85-90 %, sizable opening on west side but thermowell #2 heavily encased in cold cap.
		5:14	38.5	CC~85 unchanged.
		5:25	38.7	Still large opening on west side but ridge appears somewhat bigger. CC ~85-90%.
		5:41	38.9	$CC \sim 85\%$ , from the north viewport cap has thickened but an opening on the ridge remains letting liquid from the surface out. Lance 2 opening not visible from the middle view port. South viewport shows the existing mound only now appears smaller than previously observed. Light from the east side also observed.
		6:23	39.6	CC ~85-90%, large ridge in center of melter also large cone shape growth via south view port.
		6:50	40.1	CC ~90%, large ridge in center opening on S-E and N-W corners. Manually removed cone from below feed tube.
		7:21	40.6	Very thick hard cold cap in Thermowell 2 area. CC ~95%.
		7:27	40.7	Now observing liquid flow into Lance 2 opening, CC~90.
2	8/13/2008	7:42	41.0	CC ~90%, with liquid flowing to Lance 2 opening.
		8:00	41.3	Lance 2 area slightly more open than previously observed, feed still flowing to opening.
		8:19	41.6	CC ~90% with liquid still boiling over into Lance 2 area in between the feed shots.
		8:45	42.0	CC ~90% with liquid still boiling over into Lance area in between the feed shots.
		8:53	42.1	Very little liquid flowing into Lance 2 area.
		9:07	42.4	CC ~85%, liquid flowing to Lance 2 area better now. Some light can be seen thru center view port.
		9:55	43.2	CC ~85%, feed flowing into opening on N-W corner opening visible on S- E corner via S-W view port.
		10:10	43.4	CC ~80-85%, large ridge in center. Feed flowing into N-W corner. Opening on S-E and N-W corners.
		10:35	43.8	CC ~85%, with large ridge in center. Feed flowing into center of ridge and into melt pool thru openings in ridge. Openings in cold cap on S-E and N-W corners.
		11:15	44.5	CC ~85%, with ridge in center. Openings in S-E and NW corners with feed flowing into openings.
		11:52	45.1	CC ~80-85%, more light visible reflecting on east side. Liquid flowing to Lance 2 area.
		12:06	45.4	CC ~80-85% with feed primarily flowing to Lance 2 area.

Table 5.3. Operator	<b>Observations of Col</b>	ld Cap (CC) (continued).
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Test	Date	Time	Run Time (hours)	Cold Cap Observations				
		12:36	45.9	CC ~90%, with Lance 2 area is closing up slightly. Thermowell 2 partially encased in cap now. Cone in center area is taller.				
		12:50	46.1	CC ~85% with large ridge in center cone below feed tube. On S-E and N-W corners. Feed flowing into NW opening.				
2	9/12/2009	14:32	47.8	CC ~90%, west side has closed up around thermowell 2. No other changes.				
2	8/13/2008	14:48	48.1	CC ~90%.				
		15:04	48.3	CC ~90%, through south view port there is a large cone shape mound underneath the feed tube. West side ridge has a couple of openings allowing liquid to flow through. The ridge is dry looking.				
		15:49	49.1	CC~95%.				
		16:10	49.4	CC ~90%, part of the cone has collapsed about 4-5" off the top. The west ridge is melting. Feed is flowing into the melt pool.				
		16:25	49.7	CC~85%, west side has opened up some but only along the west wall.				
		16:38	49.9	CC ~85%, No visible changes at this time.				
		17:17	50.5	CC ~85%, ridge is melting on west side. The cone has grown some.				
	8/13/2008	17:34	50.8	CC ~85%, the west side ridge has a bigger hole in it from melting.				
		17:50	51.1	CC %85%, unchanged.				
		18:12	51.5	CC ~85%, west side is melting a lot. It looks like the ridge is about to collapse. Large amount of liquid that can be seen through the ridge.				
						18:36	51.9	Cold cap ~85%, ridge continues to melt, boiling feed flows into melter. Cone is larger.
		18:54	52.2	CC ~85%.				
		19:12	52.5	CC ~80, ridge is no longer melting down.				
3		19:31	52.8	CC ~85%, west side ridge is not melting as fast as previous.				
		19:47	53.0	CC ~85%, no visible changes.				
		20:05	53.3	CC ~80%, east side has opened up.				
		20:23	53.6	CC ~85%, west side is closing up.				
		20.42 54.0 CC ~85%, ridge appears slightly larger feed can still b	CC ~85%, ridge appears slightly larger feed can still be seen boiling and flowing into glass pool.					
		20:59	54.2	$CC \sim 80\%$ , cap is mostly the same.				
	21:17 $54.5$ CC ~80%, opening on west side a little larger, fe and flowing into glass pool. $21:40$ $54.9$ CC ~85%, part of both the ridge and the cone have							
		21:40	54.9	CC ~85%, part of both the ridge and the cone have collapsed. Large quantity of feed flows into glass pool with each shot.				
		21:57	55.2	CC ~80%, conditions unchanged.				
		22:15	55.5	CC ~85%, unchanged.				
		22:52	56.1	CC ~70%, large opening on west side of melter and very bright glow on west side of melter.				

Table 5.3. Operator Observations of Cold	Cap (CC) (continued).
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Test	Date	Time	Run Time (hours)	Cold Cap Observations
		23:11	56.4	CC ~70%, appearance unchanged.
	8/13/2008	23:23	56.6	CC ~75%.
		23:43	57.0	CC 75%, basically unchanged.
		0:05	57.3	CC $\sim$ 75%, opening on west side has closed up somewhat. Still lots of light on east side.
		0:20	57.6	CC ~75%.
		0:35	57.8	CC ~75-80%, lots of feed rushing into opening on west side of melter.
		0:55	58.2	CC ~75-80% basically unchanged.
		1:34	58.8	CC ~75-80%, large opening on west side of surface, but feed is constantly flowing into it. Large mound is visible on south side but it does not have pocket at the top to collect liquid.
		1:58	59.2	CC ~80%, heavy liquid flowing from the surface, mound at the center and light from Lance 1 opening.
		2:35	59.8	CC ~70-80%, no significant change.
3		3:02	60.3	CC $\sim$ 75-80%, still have large amount of feed flowing into opening on west side.
5		3:12	60.5	CC ~75-80%.
	8/14/2008	3:25	60.7	CC~80-85%, feed rushing into opening on west side. Mound is still in place with lots of light on east wall.
		3:40	60.9	CC ~80-85%, opening on west has feed flowing directly into it, no ridge in view. Large amount of still on east side.
		3:55	61.2	CC~80%, large opening on west side cap around it is very rigid. Large mound on the west side with lots of light on east wall.
		4:15	61.5	CC ~80%, west opening has a glazed ridge around it, visible after discharge. Large amount of feed spilling over ridge and into opening however. East side has mound and great deal of light along the wall.
		5:20	62.6	CC ~80%, observed heavy liquid overflow from the ridge on the west, light from the east exist. Mound on the center appears higher and heavy boiling around could be seen at times.
		5:40	62.9	Mound is still in place and ridge in center of melter
		5:54	63.2	CC ~80%, observed large influx of feed into west opening, so feed is still flowing over ridge.
		7:10	64.4	CC ~90%, west opening is much smaller than previous with liquid still flowing into opening. East area appears larger based on reflected light.

Table 5.3. Operator	<b>Observations of Col</b>	d Cap (CC) (continued).
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Test	Date	Time	Run Time (hours)	Cold Cap Observations
		7:20	64.6	CC~95%, west opening reduced to ~6" diameter with still flowing into opening. East side. Reflected light is reduced from previous observation.
		7:46	65.0	CC ~95%, with only small holes open on west, Lance 2 side. East appears about the same.
		8:05	65.3	CC ~95%, conditions of cap has not changed.
		8:50	66.1	Opening is large and not observing liquid flow. East side light is diminished.
		8:59	66.2	Ridge is collapsed causing pressure surge.
		9:19	66.6	CC ~90%,
		9:47	67.0	CC ~90-95%. Feed flowing to Lance 2 area freely.
		9:54	67.2	A shift on cap coverage on east is not visible.
		10:36	67.9	CC~90%, mainly unchanged in appearance. Liquid boils over "cone" ridge between the shots.
		11:28	68.7	Cold cap ~85-90%.
		11:45	69.0	CC ~90% with liquid moving into Lance 2 area again.Liquid in "cone" under feet tube boils vigorously in between the shots.
		12:35	69.8	CC ~90%, observations unchanged from previous observation.
		13:07	70.4	CC ~85%, Lance 2 area slightly opened up with undercut into shelf. Liquid boiling over into opening.
3	8/14/2008	13:24	70.7	CC ~85%, and 12-18" thick. Liquid flow is predominantly into Lance 2 opening.
		13:41	70.9	Light indicates a good opening on east side. Due to liquid flow predominantly to Lance 2 area.
		14:14	71.5	CC still the same as previous.
		14:38	71.9	CC ~85%. Cone like shape is large. West side ridge/dome has feed boiling over it. It looks like thermowell 2 is encased in feed (dry).
		14:50	72.1	CC ~85%, no visible changes.
		15:05	72.3	$CC \sim 85\%$ , very little to no change from the east. West side is able to see the top of the dome.
		16:10	73.4	CC~ 85%, perforated ridge on west allows feed top flow into melt pool in between the shots. Moderate size cone is seen trough south view port.
		16:31	73.8	CC ~85%, cone appears slightly larger, other conditions are not changed.
		16:49	74.1	CC ~85%, conditions mostly unchanged.
		17:05	74.3	CC ~85%, ridge on west side looks like it is fillings in, feed is still boiling and flowing into the melter. Cone appears unchanged.
		17:29	74.7	CC ~90%, both sides look slightly darker, also a ceiling like growth can be seen in the upper part of the north view port.
		17:45	75.0	CC ~90%, unchanged.
		18:13	75.5	CC ~85%, west side ridge seems smaller; slightly more light can be seen on the east side.
		18:32	75.8	CC ~85%, west ridge appears to be melting. Feed is boiling and flowing into melt pool, cone looks the same.

Table 5.3. Operator Observations of Cold	Cap (CC) (continued).
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Test	Date	Time	Run Time (hours)	Cold Cap Observations
		18:50	76.1	CC ~90%, west side has closed up some. Large cone underneath feed tube.
		19:18	76.6	CC ~90%, west side closing up.
		19:51	77.1	CC ~90%.
		20:20	77.6	CC ~85%. Slightly darker on east side, otherwise unchanged.
		20:36	77.9	CC ~90%, both sides slightly darker.
		20:54	78.2	CC ~90%, mostly the same.
		21:13	78.5	CC ~90%, slight darker on sides of the melter.
		21:27	78.7	CC ~90%, very little change, maybe a little darker on both sides.
	8/14/2008	21:47	79.0	CC ~90%, east side looks a little lighter.
		22:15	79.5	CC ~90-95%, west side darker and east side unchanged.
		22:47	80.0	CC ~80% very little light visible on west side. Large mound still visible in south view port.
		23:12	80.5	CC ~80, with a large mound on south east side of the melter.
		23:35	80.8	Attempt to break down mound manually.
		23:43	81.0	CC ~90%, light on the east exists. Previous attempts were able to break away part of the volcano releasing liquid accumulation.
3		23:54	81.2	West side of the melter is completely dark with no molten glass visible, there appears to be a hard dark crusty shell over melt surface. That is building quicker than its melting.
		0:23	81.6	CC ~90-95%. Some light now visible on west side but with crusty ridge above it. Almost no light visible on east side.
		0:33	81.8	No light on the east side.
		0:54	82.2	CC ~85-90 %, West side has opened up considerably. Slight amount of light on the east side.
		1:17	82.5	Large opening on the west side with glazed ridge overhanging it. Slight amount of light on east side.
	0/15/2000	1:36	82.9	CC ~85%, large opening on west side unchanged, small amount of light on east side.
	8/15/2008	1:56	83.2	CC~85% unchanged.
		2:06	83.4	CC ~85%, cold cap wall at the Lance 2 opening is glazed. Observed no liquid flowing. To this opening in between shots. Lance 1 opening exists with some light reflection; partial mound at south view port is visible.
		2:25	83.7	CC ~85%, opening at Lance 2n slightly larger. Thin layer of cap floating on the glass, glazed wall with no change.
		2:44	84.0	CC ~85%, glazed wall on the west still no liquid overflow. Light reflection on the east slightly brighter.
		3:02	84.3	CC ~85%, west section observed slowly dissolving.

Test	Date	Time	Run Time (hours)	Cold Cap Observations
		3:07	84.4	Liquid starting to overflow from the cap to the opening on the west.
		3:43	85.0	CC ~85%, west has a thin cap on the glass have now solidified about 2-3", glazed wall is still present
		4:20	85.6	CC ~85%, unchanged.
		5:25	86.7	CC ~80-85%, mound starting to build up again.
		6:05	87.3	CC ~80-85 % unchanged.
		7:08	88.4	$CC \sim 95\%$ , cone is fairly high and limiting liquid dispersion. Dislodged top portion of cone. Suspected partial full line clog, But subsequent shots were fine. Performed manual line flush.
		7:25	88.7	CC ~90-95%, liquid flowing to Lance 2 opening. An area with ~8-10" diameter is open now.
		7:50	89.1	CC ~95%, east opening appears to be closing, observe only a small amount of light.
		7:54	89.2	Believed most if not all feed is flowing to east side. No liquid flow to Lance 2 area is observable.
		8:00	89.3	Observed liquid flowing to Lance 2 area now. East side appears unchanged.
		8:07	89.4	Observed large influx of liquid to Lance 2 area.
3	8/15/2008	8:20	89.6	CC ~90%.
		9:10	90.4	CC ~ 85-90 %, light is visible from SE corner. Opening on NW corner has been flowing into melt pool.
	8/15/2008	9:50	91.1	CC ~90%, light visible on SE corner. Mound in center below feed tube. NW Corner is pen with feed flowing surface.
		10:10	91.4	CC unchanged.
		10:25	91.7	CC ~90%, Light is visible on S_W corner. Cone below feed tube is larger. Opening on NW corner has feed flowing through cold cap/into melt surface.
		11:22	92.6	Cold cap ~85%. Lance 2 area is opened up. Liquid is flowing into opening. East side appears unchanged.
		11:46	93.0	CC ~85-90% with flowing strongly into Lance 2 opening. Light visible on east appears about the same as previous.
		12:20	93.6	CC ~90%, light visible on S-E corner. Cone is still below feed tube. N-W opening has been flowing thru cold cap.
		12:38	93.9	CC unchanged.
		12:55	94.2	CC ~90%, little changed since last full description.
		13:15	94.5	CC still very stable.

Table 5.3. Operator	· Observations of	f Cold Cap	(CC) (continued).
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Test	Date	Time	Run Time (hours)	Cold Cap Observations
		13:39	94.9	Dislodged cone during discharge.
		14:32	95.8	$CC \sim 90\%$ , west side has a large ridge that has a couple of holes in it, which is allowing liquid feed to flow into the melt pool. The cone shaped mound is the same height as the south view port.
		14:48	96.1	CC ~90%, No visible changes.
		15:10	96.4	CC ~90%, west side ridge looks dry in spots and glazed in other places.
		15:23	96.6	CC ~90%, cone shaped mound has grown some. No other changes.
		15:42	97.0	CC ~90%, it seems to be stable at this point.
		10:04	91.3	CC ~90%, large ridge on the west side. Looks more like a double wall than a cone under feed tube area
		16:26	97.7	CC ~90%, most of the west side is glazed. The cone shape mound looks higher now.
		16:40	97.9	CC ~90%, the hole in west ridge is a little larger allowing liquid to flow onto the melt pool.
		16:55	98.2	CC ~90-95%. West side is closing up.
		17:15	98.5	CC ~90%, west side opened up some since discharge. Able to see more light from the east side. Cone shaped mound does not look any bigger.
	8/15/2008	17:36	98.9	CC $\sim$ 90%. Feed is not flowing into the melt pool on the east side like it was earlier.
3		17:52	99.1	$CC \sim 90\%$ , west side ridge is glazed and extends toward the south and can be seen in the south view port. Under the feed tube the mound looks the same.
		18:11	99.4	$CC \sim 90\%$ , a small hole has opened at the bottom center of the west ridge and is allowing feed to flow into the west side of the melt pool.
		18:27	99.7	CC ~90%, hole in ridge is slightly larger; a stalactite has grown under feed tube causing shots to spray slightly to the south. Moderate tapping on feed tube including during shot dislodged about half of it. The shot is not spraying now.
		18:44	100.0	CC ~90%, little to no change in conditions.
		19:05	100.3	CC ~90%, most of the liquid is flowing into east side of melt pool. West side ridge is mostly glazed.
		19:20	100.6	CC ~90%, West side ridge melting. The stalactite has grown to top of the cone.
		19:30	100.8	End of Test 3.
		22:00	103.3	Most of the cap is gone but the cone shaped mound is still present in the south view port which looks like it is being held up between the thermowells and the bubblers.
		23:23	104.6	Majority of cold cap is gone but still a lot of build up on melter walls. Middle view port is still entirely blocked.
	8/16/2008	0:37	105.9	CC material still melting off walls, almost nothing on melter surface.
	5/10/2000	1:20	106.6	Cold cap is gone.

## Table 5.4. DM1200 Tests Performed with Final HLW Bubbler Configuration and GlassTemperature of 1150°C.

Test	Feed	Glass Yield	Duration	Bubbling Rate	Glass Production Rate
Test 1	Al-Limited Waste	500 g/l	48 hrs	124 lpm	1500 kg/m <sup>2</sup> /d
Test 2	Al-Limited Waste	500 g/l	48 hrs	71 lpm	1050 kg/m²/d
Configuration Test 9A VSL-04R4800-4 [6]	AZ-101	400 g/l	145 hrs	64 lpm	1050 kg/m²/d
Configuration Test 9B VSL-04R4800-4 [6]	AZ-101	400 g/l	72 hrs	134 lpm	1400 kg/m²/d
Test 1B VSL-05R5800-1 [23]	AZ-102	340 g/l	114 hrs	65 lpm	900 kg/m²/d
Test 2B VSL-05R5800-1 [23]	C-106/AY-102, High Waste Loading	340 g/l	105 hrs	90 lpm	1050 kg/m²/d
MACT HLW 1 (400°C plenum) VSL-05R5830-1 [50]	C-106/AY-102, spiked	430 g/l	52 hrs	24 lpm	700 kg/m <sup>2</sup> /d
MACT HLW 2A (345°C plenum) VSL-05R5830-1 [50]	C-106/AY-102, spiked	430 g/l	75 hrs	9 lpm	550 kg/m²/d
MACT HLW 1-cont (400°C plenum) VSL-05R5830-1 [50]	C-106/AY-102, spiked	430 g/l	19 hrs	28 lpm	742 kg/m <sup>2</sup> /d
MACT HLW 2B (500°C plenum) VSL-05R5830-1 [50]	C-106/AY-102, spiked	430 g/l	54 hrs	43 lpm	1072 kg/m <sup>2</sup> /d

						1					,	2		
	-			Total		Steady State				Total		Steady Sta		ate
			avg	min	max	avg	min	max	avg	min	max	avg	min	max
		13" from floor E	1152	1113	1207	1149	1113	1177	1157	1131	1177	1156	1131	1174
		15.5" from floor E	1148	1110	1203	1146	1110	1173	1153	1127	1174	1153	1127	1170
		18" from floor E	1149	1113	1203	1147	1113	1171	1153	1126	1173	1153	1126	1171
	Glass	27" from floor E	1122	965	1188	1117	965	1158	1140	1076	1165	1141	1090	1165
	01035	13" from floor W	1146	1102	1194	1145	1102	1166	1145	1103	1166	1144	1103	1165
()		15.5" from floor W	1149	1106	1193	1148	1106	1175	1147	1106	1168	1147	1106	1167
TEMPERATURE (°C)		18" from floor W	1149	1101	1193	1149	1101	1179	1147	1104	1170	1146	1104	1170
ATUI		27" from floor W	1129	738	1190	1127	738	1187	1136	1035	1169	1137	1035	1169
APER		8" below ceiling	591	452	862	559	452	763	580	482	820	572	482	664
TEN	Plenum	17" below ceiling	659	518	835	647	518	782	570	382	792	556	382	778
		Exposed	710	516	986	714	516	986	563	404	772	560	404	726
		TC 1	1018	942	1063	1023	946	1063	1024	921	1085	1035	1003	1085
	Discharge	TC 2	1050	976	1080	1057	1021	1080	1053	968	1106	1062	1029	1106
	Discharge	Air Flow	316	251	342	319	251	342	316	239	345	323	259	345
		Riser	1119	986	1156	1128	1098	1156	1109	973	1162	1122	1081	1162
		East	1141	1113	1163	1141	1113	1163	1142	1097	1169	1145	1116	1169
	Electrode	West	1131	1098	1146	1134	1117	1146	1118	1062	1134	1117	1093	1130
		Bottom	1068	1054	1082	1069	1059	1082	1056	1018	1069	1057	1039	1069
	Film	Added Air	219	125	428	194	125	357	210	128	416	197	128	362
	Cooler	Outlet	454	295	630	439	295	510	410	188	610	400	188	491
	Density	y (g/cc)	2.24	2.00	2.46	2.22	2.00	2.33	2.30	2.19	2.44	2.29	2.19	2.38
Glass	Level (" fi		31.64	28.04		31.63	28.04	34.82	31.63	29.01	33.70	31.69	30.20	33.70
	Resistanc	/	0.098	0.073				0.110	0.092	0.080	0.102	0.093	0.087	0.102
	Curre	nt (A)	661	627	734	663	627	734	739	676	855	732	676	855
Electrodes		ge (V)	148	112	155	149	139	155	125	103	141	128	113	141
		:(kW)	224	170	243	225	205	241	170	129	201	175	140	201
Lance		Rate (lpm)	67.1	6.4	121.1	68.0	40.7	121.1	32.4	6.1	75.7	35.3	20.8	75.7
Bubblers		Rate (lpm)		6.1	121.1	54.5	35.2	121.1	31.4	6.0	71.0	34.4	11.9	71.0
	nce Bubblir	· · · /	125.1	13.7	183.7	123.7	92.9	183.7	65.0	13.2	113.1	70.9	42.2	113.1
-" Empty data				C										

#### Table 5.5. Measured DM1200 Melter System Parameters (8/6/2008 – 8/16/2008).

"-" Empty data field

#### Table 5.5. Measured DM1200 Melter System Parameters (8/6/2008 – 8/16/2008) (continued).

		3							
	-		Total	-	Steady State				
			avg	min	max	avg	min	max	
		13" from floor E	1180	1150	1204	1180	1150	1204	
		15.5" from floor E	1177	1146	1206	1177	1146	1206	
		18" from floor E	1177	1143	1205	1178	1143	1205	
	Glass	27" from floor E	1165	1112	1188	1166	1112	1188	
	01055	13" from floor W	1169	1119	1191	1169	1119	1191	
() ()		15.5" from floor W	1171	1115	1198	1172	1115	1198	
TEMPERATURE (°C)		18" from floor W	1171	1107	1201	1172	1107	1201	
ATU		27" from floor W	1162	1062	1201	1164	1062	1201	
MPER		8" below ceiling	646	388	821	621	388	821	
TEI	Plenum	17" below ceiling	529	186	777	482	186	734	
		Exposed	611	142	914	562	142	898	
		TC 1	1001	833	1057	999	833	1055	
	Discharge	TC 2	1031	936	1074	1030	936	1073	
	Discharge	Air Flow	304	209	340	302	209	336	
		Riser	1143	1102	1184	1143	1117	1184	
		East	1168	1144	1183	1169	1152	1183	
	Electrode	West	1140	1111	1156	1140	1111	1156	
		Bottom	1080	1061	1092	1081	1061	1092	
	Film	Added Air	197	85	329	181	85	258	
	Cooler	Outlet	413	74	546	406	74	525	
	Density	/ (g/cc)	2.33	2.22	2.39	2.33	2.24	2.39	
Glass	Level (" fi	rom floor)	31.75	30.27	33.65	31.79	30.46	33.05	
	Resistanc	e (ohms)	0.083	0.000	0.094	0.083	0.000	0.093	
	Curre	nt (A)	677	0	1112	687	0	1112	
Electrodes	Voltag	ge (V)	123	0	142	120	0	142	
	Power	· (kW)	181	0	248	174	0	248	
Lance	1	Rate (lpm)	27.3	12.7	61.0	23.7	12.7	46.0	
Bubblers	2	Rate (lpm)	25.6	14.3	65.9	23.0	14.3	50.7	
	nce Bubbli	ng (lpm)	54.1	33.4	102.7	47.8	33.4	87.5	
mpty data fi	eld								

"-" Empty data field

			1			2			3	
	Test	Avg.	Min.	Max.	Avg.	Min.	Max.	Avg.	Min.	Max.
	Pressure at Level Detector Port ("water)	-2.9	-6.6	0.6	-3.1	-7.2	-0.2	-3.1	-7.4	0.0
Melter	Pressure at Instrument Port ("water)	-3.2	-7.1	0.5	-3.4	-7.9	-0.4	-3.5	-8.1	-0.1
	Control Air Flow Rate (scfm)	39.6	9.1	83.9	49.2	14.7	78.4			53.9
Film	Cooler Differential Pressure ("water)	2.4	0.6	10.1	1.6	0.0	4.1			10.1
	ion Line Differential Pressure ("water)	3.7	1.8	8.3	3.2	1.6	6.5			6.7
	Differential Pressure ("water)	32.3	20.6	35.7	31.8	26.1	34.7			36.0
	Inlet gas pressure ("water)	-8.9	-18.4		-7.9	-12.9	-3.5			-3.5
	Outlet gas pressure ("water)	-41.2			-39.5	-44.8	-35.2			-35.8
	Downcomer Annulus Pressure (psia)	14.1	13.7	14.3	14.2	14.0	14.3			14.4
	Inlet gas Temp. (°C)	341	224	487	292	227	475			408
	Outlet gas Temp. (°C)	46.2	41.5	51.6	39.3	33.6	51.4			42.7
	C. Coil W. Inlet Temp (°C)	20.0	18.8	21.3	18.3	15.6	21.5			19.6
	C. Coil W. Outlet Temp (°C)	35.5	31.9	40.3	30.7	24.7	42.8			33.1
	Jacket W. Outlet Temp (°C)	39.4	35.7	44.3	33.6	27.6	45.1			36.3
~ ~ ~	Sump Temp. (°C)	39.2	34.7	44.8	33.1	26.8	45.8		Min.           -7.4           -8.1           13.7           0.7           1.2           26.6           -15.5           -47.8           13.9           230           35.5           15.7           26.6           30.0           28.9           179           184           175           95           77           62           42           11.8           36.8           0.8           1.4           35.8           31.3           185           0.1           0.0           0.9           36.4           0.3           72.3           79.9           22.9           1.2           2.8	36.3
SBS	Offgas Downcomer Temp @3" (°C)	264	215	381	225	180	387			298
	Offgas Downcomer Temp @28" (°C)	272	224	384	233	189	392			304
	Offgas Downcomer Temp @23" (°C)	260	216	365	225	184	376		Min.           -7.4           -8.1           13.7           0.7           1.2           26.6           -15.5           -47.8           13.9           230           35.5           15.7           26.6           30.0           28.9           179           184           175           95           77           62           42           11.8           36.8           0.8           1.4           35.8           31.3           185           0.1           0.0           0.9           36.4           0.3           72.3           79.9           22.9           1.2	295
	Offgas Downcomer Temp @38" (°C)	221	121	333	207	140	370			285
	Offgas Downcomer Temp @48" (°C)	143	85	262	150	81	347			267
	Offgas Downcomer Temp @53" (°C)	85	74	181	110	69	284			236
	Offgas Downcomer Temp @58" (°C)	61	52	73	59	46	80		42	74
	C. Coil/Jacket W. Flow Rate (gal/min)	28.7	27.8	29.7	27.0	9.7	29.3	27.8	11.8	29.2
	Recirc. pump discharge Temp (°C)	46.3	41.4	49.7	39.5	35.7	48.2	39.6	36.8	42.1
	Recirc. pump discharge Pressure (psi)	37.2	0.3	39.4	37.2	1.6	39.0	37.2	0.8	38.8
	Differential Pressure ("water)	2.5	1.5	4.3	2.6	1.7	3.4	2.3	1.4	3.0
	Inlet gas Temp. (°C)	46.2	41.8	51.3	39.3	33.8	51.8	39.4	35.8	42.8
WEOD	Outlet gas Temp. (°C)	47.4	33.1	50.4	40.7	31.6	49.8	41.4	31.3	43.6
WESP	Wet Gas Flow Rate (scfm)	231	135	290	232	179	259	224	Min.           -7.4           -8.1           13.7           0.7           1.2           26.6           -15.5           -47.8           13.9           230           35.5           15.7           26.6           30.0           28.9           179           184           175           95           77           62           42           11.8           36.8           0.8           1.4           35.8           31.3           185           0.1           0.0           0.9           36.4           0.3           72.3           79.9           22.9           1.2           2.8           25.5           26.0	248
	Voltage (kV)	31.5	0.1	34.7	30.4	0.0	32.8	30.6	0.1	31.4
	Current (mA)	16.4	0.0	19.6	14.2	0.0	18.0	14.5	Min.           -7.4           -8.1           13.7           0.7           1.2           26.6           -15.5           -47.8           13.9           230           35.5           15.7           26.6           30.0           28.9           179           184           175           95           77           62           42           11.8           36.8           0.8           1.4           35.8           31.3           185           0.1           0.0           0.9           36.4           0.3           72.3           79.9           22.9           1.2           2.8           25.5           26.0	17.2
HEME	Differential Pressure ("water)	1.4	0.9	2.3	1.5	1.1	1.9	1.5	0.9	1.8
#1	Outlet Gas Temp. (°C)	45.6	39.5	48.3	39.9	36.8	47.8	40.4	36.4	42.4
HEPA	Differential Pressure ("water)	0.4	0.3	0.6	0.4	0.3	0.5	0.6	0.3	0.8
1	Outlet Gas Temp. (°C)	74.0		77.0	73.6	58.5	75.4	74.1	72.3	75.7
	Inlet Gas Temp. (°C)	83.7	81.2	85.2	81.0	74.6	82.9	80.5	79.9	81.1
PBS	PBS Sump Temp. (°C)	28.2	26.5	30.1	25.0	22.1	28.6	25.0	22.9	26.4
	Differential Pressure ("water)	2.4	1.3	5.0	2.7	1.6	3.8	2.3	g.         Min.           1         -7.4           5         -8.1           4         13.7           0         0.7           0         1.2           4         26.6           0         -15.5           2         -47.8           2         13.9           3         230           4         35.5           3         15.7           3         26.6           5         30.0           0         28.9           2         179           3         184           3         175           4         95           3         77           62         42           8         11.8           6         36.8           2         0.8           4         35.8           4         31.3           4         185           6         0.1           5         0.0           4         36.4           5         79.9           0         22.9           4         26.0	3.2
HEME	Differential Pressure ("water)	3.6	2.7	4.4	3.5	2.6	4.1	$\begin{array}{c} 14.2\\ 303\\ 39.4\\ 18.3\\ 30.3\\ 33.5\\ 33.0\\ 232\\ 238\\ 228\\ 194\\ 133\\ 96\\ 55\\ 27.8\\ 39.6\\ 37.2\\ 2.3\\ 39.4\\ 41.4\\ 224\\ 30.6\\ 14.5\\ 1.5\\ 40.4\\ 0.6\\ 74.1\\ 80.5\\ 25.0\\ \end{array}$	2.8	4.0
#2	Inlet Gas Temp. (°C)	30.0	27.8	31.6	26.8	24.6	29.9	27.2	25.5	28.4
	Outlet Gas Temp. (°C)	30.6	28.3	32.0	27.8	25.6	32.7	27.8	-8.1 13.7 0.7 1.2 26.6 -15.5 2 -47.8 13.9 230 35.5 15.7 26.6 30.0 28.9 179 184 175 95 77 62 42 11.8 36.8 1.4 35.8 1.4 36.8 1.4 35.8 1.4 36.8 1.4 35.8 31.3 185 0.1 0.0 0.9 36.4 0.9 1.2 2.8 2.8 2.5 2.8 2.5 2.6 0.0 0.9 1.2 2.8 2.5 1.5 0.1 0.0 0.9 1.2 2.5 1.5 0.1 0.0 0.9 1.2 2.5 1.5 0.1 0.0 0.9 1.2 2.8 1.2 1.5 1.5 1.5 1.5 1.5 1.5 1.5 1.5	28.9
Exha	ust Stack Absolute Pressure ("water)	-9.8	-10.1	-9.6	-9.8	-10.0	-9.6	-9.8	-10.0	-9.6

Type of Sample	Test	Date	Sample ID for last Blow-down	Measured pH	Number of Blow-downs per test	Blow-down per Test (gal)	
	1	8/8/2008	S-12J-106A	2.20	29	1176.49	
SBS	2	8/13/2008	S-12K-46A	2.13	21	853.84	
	3	8/15/2008	S-12K-129A	1.97	24	952.51	
	1	8/8/2008	W-12J-110A	2.67	4	196 51	
	1	8/8/2008	W-12J-110B	3.17	4	186.51	
WESP	2	0/12/2009	W-12K-46A	2.86	4	209.83	
WESP	2	9/12/2008	W-12K-46B	3.46	4		
	3	8/16/2008	W-12K-138A	2.71	4	213.83	
	3	8/16/2008	W-12K-138B	4.37	4	213.03	
	1	8/8/2008	P-12J-110A	8.70	8	266.62	
PBS	2	8/13/2008	P-12K-47A	8.89	5	159.23	
	3	8/15/2008	P-12K-138A	8.94	4	155.08	
	1	8/8/2008	H1-12J-108A	2.63	1	16.16	
HEME 1	2	8/13/2008	H1-12K-46A	2.68	1	20.10	
	3	8/16/08	H1-12K-138A	2.68	1	14.63	
	1	8/8/2008	H2-12J-108A	7.56	2	16.97	
HEME 2	2	8/13/2008	H2-12K-46A	7.46	1	4.75	
	3	8/16/08	H2-12K-138A	8.13	2	32.36	

Table 5.7. Off-Gas Solution Volumes.

Sample Type		SBS		WESP							
Sample I.D.	S	-12K-129	A	W	/-12K-138	A	W-12K-138B				
Deluge		No	_	Be	fore Delu	uge	A	ge			
Fraction	Sus.	Dis.	Total	Sus.	Dis.	Total	Sus.	Dis.	Total		
Solids	280	3122	3402	6	1491	1497	96	498	594		
Al	8.86	191.78	200.64	< 0.01	1.70	1.70	0.18	0.34	0.52		
В	0.33	571.46	571.79	0.01	104.8	104.81	0.55	13.42	13.97		
Ba	< 0.01	1.44	1.44	< 0.01	0.07	0.07	< 0.01	0.04	0.04		
Bi	7.69	13.99	21.68	0.05	2.44	2.49	17.92	< 0.01	17.92		
Ca	2.01	53.72	55.74	< 0.01	48.83	48.83	0.09	44.05	44.14		
Cd	< 0.01	2.08	2.08	< 0.01	1.68	1.68	0.46	0.51	0.97		
Cu	2.76	11.25	14.01	0.03	25.35	25.38	11.70	1.20	12.90		
Fe	47.64	58.08	105.72	0.03	3.25	3.28	5.67	< 0.01	5.67		
K	0.17	13.91	14.08	0.01	20.9	20.91	0.27	7.28	7.55		
Li	0.33	34.45	34.78	0.01	32.66	32.67	0.09	8.57	8.66		
Mg	0.08	7.83	7.91	< 0.01	11.99	11.99	< 0.01	12.25	12.25		
Mn	0.33	2.53	2.86	< 0.01	0.08	0.08	< 0.01	< 0.01	< 0.01		
Na	0.25	236.14	236.39	< 0.01	214.35	214.35	< 0.01	59.02	59.02		
Ni	2.26	4.65	6.91	0.01	0.34	0.35	0.46	0.09	0.55		
Р	1.17	8.82	9.99	< 0.01	1.24	1.24	< 0.01	< 0.01	< 0.01		
Pb	1.50	16.23	17.73	0.04	4.67	4.71	12.43	0.77	13.20		
Si	54.16	80.97	135.13	0.02	7.1	7.10	1.28	3.25	4.53		
Sr	< 0.01	1.94	1.94	< 0.01	0.43	0.43	< 0.01	0.32	0.32		
Zn	0.50	5.82	6.32	< 0.01	0.54	0.54	0.46	0.06	0.52		
Zr	2.09	6.10	8.19	< 0.01	0.19	0.19	< 0.01	< 0.01	< 0.01		
Cl	NA	1284.9	NC	NA	194.88	NC	NA	25.39	NC		
F	NA	92.52	NC	NA	45.19	NC	NA	34.05	NC		
Nitrate	NA	199.49	NC	NA	76.98	NC	NA	18.16	NC		
Nitrite	NA	< 0.1	NC	NA	0.17	NC	NA	<0.1	NC		
Sulfate	NA	647.95	NC	NA	314.98	NC	NA	71.49	NC		

### Table 5.8. Analytical Results for Solutions from the DM1200 Off-gas System Sampled at End of Testing (mg/L).

NA – Not analyzed; NC - Not calculated

Sample Type		PBS		HE	ME	
Sample I.D.	P-12	2K-138A	H1-12	2K-138A	H2-12	2K-138A
Fraction	Sus.	Dis.	Sus.	Dis.	Sus.	Dis.
Solids	40	9859	<1	406	<1	1252
Al	NA	2.64	NA	1.36	NA	0.23
В	NA	5.12	NA	14.52	NA	1.17
Ba	NA	< 0.01	NA	0.06	NA	0.06
Bi	NA	< 0.25	NA	< 0.25	NA	< 0.25
Ca	NA	2.45	NA	46.87	NA	63.36
Cd	NA	< 0.03	NA	0.14	NA	< 0.03
Cu	NA	< 0.01	NA	0.65	NA	0.01
Fe	NA	< 0.05	NA	2.18	NA	0.06
K	NA	12.39	NA	4.21	NA	14.34
Li	NA	0.08	NA	0.54	NA	0.64
Mg	NA	0.91	NA	11.85	NA	15.76
Mn	NA	< 0.04	NA	0.09	NA	0.06
Na	NA	3305.8	NA	18.98	NA	226.13
Ni	NA	< 0.04	NA	0.18	NA	0.23
Р	NA	<0.6	NA	0.96	NA	0.92
Pb	NA	< 0.1	NA	0.69	NA	< 0.1
Si	NA	1.05	NA	3.85	NA	2.82
Sr	NA	0.04	NA	0.38	NA	0.50
Zn	NA	0.09	NA	0.59	NA	0.61
Zr	NA	< 0.02	NA	0.11	NA	< 0.02
Cl	NA	15.56	NA	33.81	NA	2.17
F	NA	91.06	NA	37.37	NA	44.58
Nitrate	NA	84.73	NA	200.26	NA	506.85
Nitrite	NA	2839.77	NA	1.91	NA	227.94
Sulfate	NA	313.97	NA	136.81	NA	100.23

# Table 5.8. Analytical Results for Solutions from the DM1200 Off-gas System Sampled at end of Testing (mg/L) (continued).

NA - Not analyzed

Formulation	Melter	Test	Al Source	Data	Nomo	%	Density	nII	Glass Yield	Glass Yi	eld (kg/kg)	%Dev. From			
Formulation	Туре	Test	AI Source	Date	Name	Water	(g/ml)	pН	(g/l)	Target	Measured	Target			
				1/29/08	BLX-F-50B	55.81	1.41	10.36	507	0.358	0.360	0.53			
		1		2/4/08	BLX-F-62A	NA	1.38	10.41	NA	0.358	NA	NC			
HLW-E-Al-		1	Al(OH) <sub>3</sub>	2/5/08	BLX-F-71A	NA	1.37	10.46	NA	0.358	NA	NC			
27			AI(011)3	2/3/08	BLX-F-79A	NA	1.38	10.17	NA	0.358	NA	NC			
		2		2/6/08	BLX-F-91A	NA	1.38	10.18	NA	0.358	NA	NC			
		4		2/0/08	BLX-F-104A	NA	1.38	10.20	NA	0.358	NA	NC			
					BLX-F-125A	58.43	1.37	8.12	487	0.360	0.356	-1.11			
		3	3		5/5/08	BLX-F-138A	NA	1.37	8.18	NA	0.360	NA	NC		
				3	Al(OH) <sub>3</sub>		BLX-F-145A	NA	1.39	8.12	NA	0.360	NA	NC	
			AI(OII) <sub>3</sub>	5/6/08	BLX-F-153A	NA	1.38	8.13	NA	0.360	NA	NC			
	DM			5/7/08	BLY-F-15A	NA	1.38	8.28	NA	0.360	NA	NC			
HWI-Al -16	100			5/8/08	BLY-F-22A	NA	1.37	8.24	NA	0.360	NA	NC			
11 w 1-A1 -10	100			6/2/08	BLY-F-57A	60.00	1.36	8.34	487	0.363	0.358	-1.32			
		5	41.0	0/2/08	BLY-F-59A	NA	1.38	8.33	NA	0.363	NA	NC			
		3		41.0	41.0	A1.0	A1 O	6/3/08	BLY-F-66A	NA	1.38	8.39	NA	0.363	NA
			Al <sub>2</sub> O <sub>3</sub>	6/4/08	BLY-F-74A	NA	1.37	8.42	NA	0.363	NA	NC			
		6		0/4/08	BLY-F-84A	NA	1.37	8.61	NA	0.363	NA	NC			
		6		6/5/08	BLY-F-99A	NA	1.37	8.44	NA	0.363	NA	NC			
				6/23/08	BLY-F-132A	57.60	1.38	8.10	479	0.361	0.347	-3.93			
				0/23/08	BLY-F-133A	NA	1.37	8.16	NA	0.361	NA	NC			
HWI-Al-19		7	Al(OH) <sub>3</sub>	6/21/09	BLY-F-141A	56.44	1.39	8.06	491	0.361	0.353	-2.22			
				$(1)^{12} = 6/24/08 =$	BLY-F-147A	NA	1.38	8.15	NA	0.361	NA	NC			
				6/25/08	BLZ-F-10A	NA	1.40	8.21	NA	0.361	NA	NC			

#### Table 6.1. Characteristics of Melter Feed Samples.

NA – Not analyzed; NC – Not calculated

								-				
Formulation	Melter	Test	Al Source	Date	Name	%	Density	pН	Glass Yield	Glass Yi	eld (kg/kg)	%Dev. From
ronnulation	Type	Test	Al Source	Date	Indiffe	Water	(g/ml)	pm	(g/l)	Target	Measured	Target
		8		6/26/08	BLZ-F-18A	NA	1.39	8.04	NA	0.361	NA	NC
	DM100	o		6/27/08	BLZ-F-28A	NA	1.40	8.11	NA	0.361	NA	NC
				8/6/08	F-12J-44A	55.40	1.38	8.17	497	0.361	0.360	-0.17
		1		8/7/08	F-12J-79A	56.02	1.39	8.10	480	0.361	0.345	-4.32
HWI-Al-19			Al(OH) <sub>3</sub>	8/8/08	F-12J-106A	58.90	1.36	8.46	446	0.361	0.328	-9.09
ПWI-АІ-19	DM	2	AI(OII)3	8/11/08	F-12J-135A	56.97	1.38	8.42	469	0.361	0.340	-5.84
	1200	4		8/12/08	F-12K-7A	56.50	1.39	8.12	477	0.361	0.343	-4.93
				8/13/08	F-12K-47A	58.82	1.35	8.12	432	0.361	0.320	-11.36
		3		8/14/08	F-12K-69A	55.88	1.39	8.17	468	0.361	0.337	-6.65
				8/15/08	F-12K-113A	58.78	1.37	8.45	454	0.361	0.331	-8.28

#### Table 6.1. Characteristics of Melter Feed Samples (continued).

NA – Not analyzed; NC – Not calculated

 Table 6.2. Rheological Properties of Melter Feed Samples.

Test	Formulation	Al Source	Date	Name	%	Density		Yield Stress	Vis	scosity (Po	oise)
Test	Formulation	AI Source	Date	Iname	Water	(g/ml)	рН	(Pa)	@1/s	@10/s	@100/s
[2]	HLW-E-Al-27	$Al_2O_3$	10/2/06	BLN-F-9A	61.23	1.37	10.90	0.1	1.39	0.27	0.06
3	HWI-Al -16	Al(OH) <sub>3</sub>	5/5/08	BLX-F-125A	58.43	1.37	8.12	2.5	3.04	0.58	0.27
5	HWI-Al -16	$Al_2O_3$	6/2/08	BLY-F-57A	60.00	1.36	8.34	1.9	5.36	0.66	0.16
7	HWI-Al-19	Al(OH) <sub>3</sub>	6/24/08	BLY-F-141A	56.44	1.39	8.06	1.1	2.72	0.39	0.14

Melter Type				D	M100			
Formulation	Н	LW-E-Al-2	27			HWI-Al -16		
Al Source	Alum	inum Hydı	oxide		Aluminum	Hydroxide	Aluminu	m Oxide
Test		1	l	Torgot		3	5	
Sample I.D.	Target	BLX-	F-50B	Target	BLX-F	F-125A	BLY-	F-57A
Constituent		XRF	%Dev.		XRF	%Dev.	XRF	%Dev.
Al <sub>2</sub> O <sub>3</sub>	23.97	23.01	-4.01	23.25	22.04	-5.22	22.11	-4.91
B <sub>2</sub> O <sub>3</sub> *	15.19	15.19	NC	17.73	17.73	NC	17.73	NC
BaO	0.05	0.07	NC	0.05	0.05	NC	0.05	NC
Bi <sub>2</sub> O <sub>3</sub>	1.14	1.20	4.88	1.11	1.16	4.87	1.26	13.56
CaO	6.08	6.03	-0.71	5.89	6.04	2.53	5.85	-0.71
CdO	0.02	0.03	NC	0.02	0.03	NC	0.02	NC
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.60	NC	0.51	0.60	NC	0.59	NC
F*	0.67	0.67	NC	0.65	0.65	NC	0.65	NC
Fe <sub>2</sub> O <sub>3</sub>	5.90	5.90	0.08	5.72	6.17	7.80	6.83	19.44
K <sub>2</sub> O	0.14	0.18	NC	0.14	0.17	NC	0.16	NC
Li <sub>2</sub> O*	3.57	3.57	NC	3.46	3.46	NC	3.46	NC
MgO	0.12	0.26	NC	0.11	0.35	NC	0.26	NC
MnO	§	0.02	NC	§	0.04	NC	0.03	NC
Na <sub>2</sub> O	9.58	9.21	-3.88	9.29	8.96	-3.59	9.06	-2.53
Nd <sub>2</sub> O <sub>3</sub>	§	< 0.01	NC	§	< 0.01	NC	< 0.01	NC
NiO	0.40	0.39	NC	0.39	0.40	NC	0.37	NC
P <sub>2</sub> O5	1.05	1.22	16.01	1.02	1.13	10.32	1.10	7.52
PbO	0.41	0.37	NC	0.40	0.37	NC	0.34	NC
SiO <sub>2</sub>	30.50	31.28	2.56	29.58	29.86	0.95	28.98	-2.02
SO <sub>3</sub>	0.20	0.23	NC	0.19	0.15	NC	0.16	NC
SrO	§	0.05	NC	§	0.02	NC	0.01	NC
TiO <sub>2</sub>	0.01	0.04	NC	0.01	0.05	NC	0.04	NC
ZnO	0.08	0.08	NC	0.08	0.09	NC	0.08	NC
ZrO <sub>2</sub>	0.39	0.39	NC	0.38	0.49	NC	0.83	NC
Sum	100.00	100.00	NC	100.00	100.00	NC	100.00	NC

#### Table 6.3. XRF Analyzed Compositions of Vitrified Melter Feed Samples (wt%).

§ - Not a target constituent\* - Target values

NC – Not calculated

Table 6.3. XRF Analyzed Compositions of Vitrified Melter Feed Samples (wt%)
(continued).

Melter Type			DM100				DM	1200	
Formulation		I	HWI-Al-19	9			HWI	Al-19	
Al Source		Alumi	num Hyd	roxide		А	Aluminum Hydroxide		
Test			7				1		2
Sample I.D.	Target	BLY-F	F-132A	BLY-F	F-141A	F-12J- 44A	F-12J- 79A	F-12J- 106A	F-12J- 135A
Constituent		XRF	%Dev.	XRF	%Dev.	XRF	XRF	XRF	XRF
Al <sub>2</sub> O <sub>3</sub>	23.97	22.88	-4.56	22.73	-5.18	22.61	22.74	22.69	23.18
B <sub>2</sub> O <sub>3</sub> *	19.19	19.19	NC	19.19	NC	19.19	19.19	19.19	19.19
BaO	0.05	0.06	NC	0.06	NC	0.05	0.05	0.06	0.06
Bi <sub>2</sub> O <sub>3</sub>	1.14	1.31	14.50	1.29	12.81	1.23	1.19	1.10	1.23
CaO	5.58	5.81	4.17	5.97	7.01	5.61	5.62	5.56	5.80
CdO	0.02	0.02	NC	0.02	NC	0.02	0.02	0.02	0.03
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.68	NC	0.67	NC	0.65	0.64	0.64	0.67
F*	0.67	0.67	NC	0.67	NC	0.67	0.67	0.67	0.67
Fe <sub>2</sub> O <sub>3</sub>	5.90	6.41	8.71	6.50	10.11	6.26	6.22	6.19	6.49
K <sub>2</sub> O	0.14	0.19	NC	0.20	NC	0.19	0.19	0.20	0.20
Li <sub>2</sub> O*	3.57	3.57	NC	3.57	NC	3.57	3.57	3.57	3.57
MgO	0.12	0.27	NC	0.27	NC	0.30	0.27	0.29	0.27
MnO	Ş	0.02	NC	0.02	NC	0.05	0.02	0.02	0.02
Na <sub>2</sub> O	9.58	9.01	-5.99	9.20	-3.96	9.68	9.64	10.48	9.31
Nd <sub>2</sub> O <sub>3</sub>	Ş	0.01	NC	< 0.01	NC	< 0.01	< 0.01	< 0.01	< 0.01
NiO	0.40	0.44	NC	0.45	NC	0.40	0.40	0.39	0.41
P <sub>2</sub> O5	1.05	0.90	-14.67	0.90	-14.88	1.06	1.11	1.01	1.03
PbO	0.41	0.37	NC	0.37	NC	0.35	0.34	0.31	0.34
SiO <sub>2</sub>	27.00	27.27	1.03	27.06	0.26	27.24	27.30	26.84	26.69
SO3	0.20	0.17	NC	0.16	NC	0.22	0.19	0.19	0.20
SrO	§	0.01	NC	0.01	NC	0.01	< 0.01	< 0.01	< 0.01
TiO <sub>2</sub>	0.01	0.04	NC	0.05	NC	0.05	0.04	0.04	0.05
ZnO	0.08	0.09	NC	0.10	NC	0.09	0.09	0.08	0.09
ZrO <sub>2</sub>	0.39	0.60	NC	0.56	NC	0.52	0.48	0.46	0.52
Sum	100.00	100.00	NC	100.00	NC	100.00	100.00	100.00	100.00

§ - Not a target constituent
 \* - Target values
 NC – Not calculated

Table 6.3. XRF Analyzed Compositions of Vitrified Melter Feed Samples (wt%)
(continued).

Melter Type				DM1200			
Formulation				HWI-Al-1	9		
Al Source			Alu	minum Hyd	roxide		
Test		2		3		A	
Sample I.D.	Target	F-12K-7A	F-12K-47A	F-12K-69A	F-12K-113A	Avg. for DM1200	%Dev.
Constituent		XRF	XRF	XRF	XRF	DIVI1200	
Al <sub>2</sub> O <sub>3</sub>	23.97	23.04	23.38	24.06	23.84	23.19	-3.25
B <sub>2</sub> O <sub>3</sub> *	19.19	19.19	19.19	19.19	19.19	19.19	NC
BaO	0.05	0.07	0.07	0.07	0.06	0.06	NC
Bi <sub>2</sub> O <sub>3</sub>	1.14	1.28	1.22	1.18	1.16	1.20	4.46
CaO	5.58	5.82	5.63	5.50	5.48	5.63	0.93
CdO	0.02	0.03	0.03	< 0.01	0.02	0.02	NC
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.66	0.69	0.62	0.67	0.66	NC
F*	0.67	0.67	0.67	0.67	0.67	0.67	NC
Fe <sub>2</sub> O <sub>3</sub>	5.90	6.56	6.54	6.20	6.57	6.38	8.12
K <sub>2</sub> O	0.14	0.20	0.22	0.19	0.20	0.20	NC
Li <sub>2</sub> O*	3.57	3.57	3.57	3.57	3.57	3.57	NC
MgO	0.12	0.25	0.31	0.26	0.29	0.28	NC
MnO	§	0.02	0.02	0.02	0.03	0.03	NC
Na <sub>2</sub> O	9.58	9.08	9.35	9.31	9.86	9.59	0.08
Nd <sub>2</sub> O <sub>3</sub>	§	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	NC
NiO	0.40	0.43	0.42	0.39	0.42	0.41	NC
P <sub>2</sub> O5	1.05	1.10	1.20	1.12	1.13	1.09	3.80
PbO	0.41	0.35	0.31	0.33	0.33	0.33	NC
SiO <sub>2</sub>	27.00	26.80	26.38	26.51	25.70	26.68	-1.16
SO <sub>3</sub>	0.20	0.19	0.22	0.21	0.21	0.21	NC
SrO	§	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	NC
TiO <sub>2</sub>	0.01	0.05	0.05	0.04	0.04	0.05	NC
ZnO	0.08	0.09	0.08	0.08	0.08	0.08	NC
ZrO <sub>2</sub>	0.39	0.54	0.47	0.49	0.47	0.49	NC
Sum	100.00	100.00	100.00	100.00	100.00	100.00	NC

§ - Not a target constituent\* - Target values

NC – Not calculated

				Tests.					
Test	T(°C)	Date	Name	Analysis	Mass (kg)	Cumulative Mass (kg)			
		2/4/08	BLX-G-65A	-	-	-			
		2/4/00	BLX-G-67A	XRF, F	28.72	28.72			
			BLX-G-67B	-	-	-			
			BLX-G-71A	XRF	33.80	62.52			
			BLX-G-71B	-	-	-			
			BLX-G-71C	XRF	31.34	93.86			
		2/5/08	BLX-G-76A	-	-	-			
		2/3/08	BLX-G-76B	XRF	23.34	117.20			
1	1200		BLX-G-77A	-	-	-			
			BLX-G-79A	XRF, F	26.06	143.26			
			BLX-G-79B	-	-	_			
			BLX-G-82A	XRF	26.12	169.38			
			BLX-G-82B	-	-	-			
			BLX-G-84A	XRF	27.32	196.70			
			BLX-G-84B	-	-	-			
		<b>2</b> 16 10 0	BLX-G-84C	XRF, PCT, TCLP	25.48	222.18			
		2/6/08	BLX-G-89A	-	-	-			
			BLX-G-91A	XRF, F	33.50	255.68			
			BLX-G-93A	-	-	-			
			BLX-G-96A	XRF	21.40	277.08			
			BLX-G-98A	_	-	_			
			BLX-G-99A	XRF	23.10	300.18			
		2/7/08	BLX-G-102A	-	-	-			
			BLX-G-102B	XRF	23.76	323.94			
							BLX-G-104A	-	-
2	1150		BLX-G-108A	XRF, F	20.62	344.56			
-			BLX-G-109A	-	-	-			
			BLX-G-113A	XRF	31.96	376.52			
		2/8/08	BLX-G-114A	-	-	-			
		2/0/00	BLX-G-114B	XRF	24.24	400.76			
			BLX-G-114C	-	-	-			
			BLX-G-115A	XRF,PCT, TCLP	21.00	421.76			
			BLX-G-119A	XRF	35.78	457.54			
		2/9/08	BLX-G-119B	XRF, F	32.56	490.10			
		5/5/08	BLX-G-143A	-	52.50	-			
		0,0,00	BLX-G-144A	XRF, F	20.16	510.26			
			BLX-G-144A BLX-G-144B	-	20.10	-			
			BLX-G-144B BLX-G-145A	XRF	24.26	534.52			
			BLX-G-145A BLX-G-145B	-	-	-			
3	3 1200	5/6/08	BLX-G-143B BLX-G-147A	XRF	24.72	559.24			
		5/0/00	BLX-G-14/A BLX-G-151A	-	-	-			
			BLX-G-151A BLX-G-151B	XRF	28.90	588.14			
			BLX-G-151B BLX-G-152A	- ARF		-			
					25.02	613.16			
	1	a fiald	BLX-G-153A	XRF, F	23.02	013.10			

#### Table 6.4. Listing of Glass Discharged, Masses, and Analysis Performed During DM100 Tests.

Test	T(°C)	Date	Name	Analysis	Mass (kg)	Cumulative Mass (kg)
		5/6/08	BLX-G-154A	-	-	-
		5/0/08	BLY-G-8A	XRF	22.98	636.14
			BLY-G-8B	-	-	-
3	1200		BLY-G-10A	XRF	23.50	659.64
3	1200		BLY-G-10B	-	-	-
			BLY-G-10C	XRF	23.50	683.14
		5/7/08	BLY-G-14A	-	-	-
		5/7/08	BLY-G-14B	XRF,PCT,TCLP,DCP,F	31.88	715.02
			BLY-G-15A	-	-	-
			BLY-G-16A	XRF	21.78	736.80
			BLY-G-16B	-	-	-
			BLY-G-17A	XRF	19.68	756.48
			BLY-G-17B	-	-	-
			BLY-G-22A	XRF	21.20	777.68
			BLY-G-22B	-	-	-
			BLY-G-25A	XRF, F	18.66	796.34
		5/8/08	BLY-G-25B	-	-	-
		5/8/08	BLY-G-26A	XRF	23.14	819.48
		BLY-G-26B	-	-	_	
4	1150		BLY-G-26C	XRF	18.84	838.32
			BLY-G-28A	-	-	_
			BLY-G-28B	XRF	18.64	856.96
			BLY-G-32A	-	-	-
			BLY-G-34A	XRF, F	18.64	875.60
			BLY-G-34B	-	-	-
			BLY-G-35A	XRF	20.78	896.38
		5/9/08	BLY-G-36A	-	-	_
			BLY-G-36B	XRF,PCT,TCLP,DCP	24.28	920.66
			BLY-G-40A	XRF	28.42	949.08
			BLY-G-40B	XRF, F	29.14	978.22
			BLY-G-40C	XRF	17.72	995.94
			BLY-G-62A	-	-	-
			BLY-G-63A	XRF, F	28.70	1024.64
			BLY-G-66A	-	-	-
			BLY-G-66B	XRF	32.86	1057.50
		6/3/08	BLY-G-68A	-	-	-
-	1000		BLY-G-69A	XRF	26.66	1084.16
5	1200		BLY-G-69B	-	-	-
			BLY-G-73A	XRF	26.26	1110.42
			BLY-G-74A	-	-	-
			BLY-G-75A	XRF, F	28.64	1139.06
		6/4/08	BLY-G-79A	-	-	-
			BLY-G-80A	XRF	27.14	1166.20

### Table 6.4. List of Glass Discharged, Masses, and Analysis Performed During DM100 Tests (continued).

Test	T(°C)	Date	Name	Analysis	Mass (kg)	Cumulative Mass (kg)
			BLY-G-80B	-	-	-
5	1200		BLY-G-80C	XRF,PCT,TCLP,DCP	27.40	1193.60
		6/4/08	BLY-G-84A	_	-	-
			BLY-G-85A	XRF	33.64	1227.24
			BLY-G-87A	-	-	-
			BLY-G-91A	XRF, F	22.68	1249.92
			BLY-G-92A	-	-	-
			BLY-G-93A	XRF	20.42	1270.34
			BLY-G-93B	-	-	-
		6/5/08	BLY-G-93C	XRF	15.78	1286.12
			BLY-G-94A	-	-	-
			BLY-G-98A	-	-	-
6 1150	1150		BLY-G-98B	XRF	20.34	1306.46
6	1150		BLY-G-99A	-	-	-
			BLY-G-101A	XRF	15.56	1322.02
			BLY-G-106A	-	-	-
		()()00	BLY-G-107A	XRF, F	28.62	1350.64
	6/6/08	BLY-G-111A	-	-	_	
			BLY-G-111B	XRF	23.54	1374.18
			BLY-G-112A	-	-	-
		6/7/08	BLY-G-114A	XRF,PCT,TCLP,DCP	21.76	1395.94
		6/10/00	BLY-G-120A	XRF	27.16	1423.10
		6/19/08	BLY-G-120B	XRF	41.04	1464.14
		6/23/08	BLY-G-135A	-	-	-
			BLY-G-139A	XRF, F	32.24	1496.38
			BLY-G-141A	-	-	-
			BLY-G-141B	XRF	33.58	1529.96
			BLY-G-144A	-	-	-
		6/24/08	BLY-G-144B	XRF	30.16	1560.12
			BLY-G-145A	-	-	-
_	1.000		BLY-G-145B	XRF	31.38	1591.50
7	1200		BLY-G-147A	-	-	-
			BLY-G-151A	XRF, F	33.98	1625.48
			BLY-G-151B	-	_	-
			BLY-G-152A	XRF	30.34	1655.82
			BLY-G-153A	_	-	_
			BLY-G-153B	XRF	26.16	1681.98
		6/25/00	BLZ-G-5A		-	-
			BLZ-G-5B	XRF,PCT,TCLP,DCP	27.36	1709.34
			BLZ-G-11A	-	_	-
			BLZ-G-13A	XRF, F	31.96	1741.30
8	1150		BLZ-G-13B	,-	-	-
		6/26/08	BLZ-G-16A	XRF	25.74	1767.04

#### Table 6.4. List of Glass Discharged, Masses, and Analysis Performed During DM100 Tests (continued).

### Table 6.4. List of Glass Discharged, Masses, and Analysis Performed During DM100 Tests (continued).

Test	T(°C)	Date	Name	Analysis	Mass (kg)	Cumulative Mass (kg)
			BLZ-G-18A	-	-	-
			BLZ-G-21A	XRF	24.62	1791.66
		6/26/08	BLZ-G-21B	-	-	-
		0/20/08	BLZ-G-23A	XRF	24.32	1815.98
			BLZ-G-23B	-	-	-
			BLZ-G-27A	XRF, F	25.28	1841.26
			BLZ-G-27B	-	-	-
8	1150		BLZ-G-28A	XRF	25.70	1866.96
			BLZ-G-29A	-	-	-
			BLZ-G-29B	XRF	22.60	1889.56
		6/27/08	BLZ-G-33A	-	_	-
			BLZ-G-33B	XRF	23.06	1912.62
			BLZ-G-35A	-	-	-
			BLZ-G-35B	XRF, F	24.14	1936.76
			BLZ-G-37A	XRF, PCT, TCLP, DCP	11.64	1948.40

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Гest	Date	Name	Analysis	Mass (kg)	Cumulative Mass (kg)
		G-12J-41A	-	-	-
		G-12J-42A	-	-	-
		G-12J-42B	-	-	-
		G-12J-43A	-	-	-
		G-12J-43B	-	-	-
	8/6/08	G-12J-44A	-	-	-
		G-12J-45A	XRF, DCP, F	478.5	478.5
		G-12J-45B	-	-	-
		G-12J-46A	-	-	_
		G-12J-48A	-	-	_
		G-12J-48B	-	-	-
		G-12J-60A		502.0	980.5
		G-12J-60B	-	-	-
		G-12J-61A	-	- 1	-
		G-12J-61B	-	-	-
		G-12J-62A	-	-	-
		G-12J-62B	-	-	-
		G-12J-64A	XRF	481.5	1462.0
		G-12J-65A	-	-	-
		G-12J-69A	_	_	-
		G-12J-69B	_	_	_
1		G-12J-70A	_	_	_
		G-12J-71A	_	_	-
		G-12J-71B	_	_	_
		G-12J-71C	_	_	_
	8/7/08	G-12J-72A	XRF, F	499.0	1961.0
		G-12J-72B	,-	-	-
		G-12J-73A	_	_	_
		G-12J-75A	_	_	-
		G-12J-75B	_	_	_
		G-12J-77A	_	-	-
		G-12J-79A	_	_	-
		G-12J-79B	XRF	485.0	2446.0
		G-12J-88A	-	-	-
		G-12J-89A	_	-	_
		G-12J-89B	_	_	_
		G-12J-90A	-	<u> </u>	-
		G-12J-90B	_	_	-
		G-12J-100A	XRF	446.5	2892.5
		G-12J-100A G-12J-100B	-	-	-
	8/8/08	G-12J-100D		<u> </u>	
	0,0,00	G-12J-101A G-12J-102A	_	-	-
		G-12J-102A G-12J-104A			-

#### Table 6.5. List of Glass Discharged, Masses, and Analysis Performed during DM1200 Test.

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Test	Date	Name	Analysis	Mass (kg)	Cumulative Mass (kg)
		G-12J-105A	-	-	-
		G-12J-105B	XRF, DCP, F	480.5	3373.0
1	8/8/08	G-12J-106A	-	-	-
		G-12J-110A	-	-	-
		G-12J-110B	-	-	-
		G-12J-133A	-	-	-
	0/11/00	G-12J-135A	XRF	492.5	3865.5
	8/11/08	G-12J-135B	-	-	-
		G-12J-136A	-	-	-
		G-12J-137A	-	-	-
		G-12J-139A	-	-	-
		G-12J-139B	-	-	-
		G-12J-148A	-	-	-
		G-12J-149A	-	-	-
		G-12J-149B	XRF, DCP	498.5	4364.0
		G-12J-152A	-	-	-
		G-12J-152B	-	-	-
	0/10/00	G-12J-154A	-	-	-
•	8/12/08	G-12J-155A	-	-	-
2		G-12K-7A	-	-	-
		G-12K-11A	-	-	-
		G-12K-11B	-	-	-
		G-12K-21A	XRF	495.5	4859.5
		G-12K-22A	-	-	-
		G-12K-23A	-	-	-
		G-12K-24A	-	-	-
		G-12K-24B	-	-	-
		G-12K-26A	-	-	-
		G-12K-26B	-	-	-
		G-12K-28A	-	-	-
		G-12K-28B	-	- 1	-
		G-12K-28C	XRF, F	498.0	5357.5
		G-12K-37A	-	- 1	-
		G-12K-37B	-	-	-
	8/13/08	G-12K-38A	-	-	-
		G-12K-38B	-	- 1	-
		G-12K-42A	-	-	-
3		G-12K-43A	-	-	-
		G-12K-43B	-	-	-
		G-12K-44A	XRF	475.0	5832.5
		G-12K-46A	-	-	-
		G-12K-47A	_	_	_

## Table 6.5. List of Glass Discharged, Masses, and Analysis Performed during DM1200 Test(continued).

Test	Date	Name	Analysis	Mass (kg)	Cumulative Mass (kg)	
		G-12K-57A	-	-	-	
		G-12K-58A	-	-	-	
	9/12/09	G-12K-58B	-	-	-	
	8/13/08	G-12K-59A	XRF	462.5	6295.0	
		G-12K-61A	-	-	-	
		G-12K-63A	-	-	-	
		G-12K-64A	-	-	-	
		G-12K-66A	-	-	-	
		G-12K-67A	-	-	-	
		G-12K-68A	-	-	-	
		G-12K-81A	-	-	-	
		G-12K-81B	XRF	513.0	6808.0	
		G-12K-84A	-	-	-	
	8/14/08	G-12K-85A	-	-	-	
	8/14/08	G-12K-86A	-	-	-	
		G-12K-87A	-	-	-	
		G-12K-87B	-	-	-	
		G-12K-89A	-	-	-	
		G-12K-91A	XRF, F	502.0	7310.0	
		G-12K-92A	-	-	-	
		G-12K-92B	-	-	-	
3		G-12K-102A	-	-	-	
		G-12K-104A	-	-	-	
		G-12K-105A	-	-	-	
		G-12K-105B	-	-	-	
		G-12K-107A	-	-	-	
		G-12K-107B	-	-	-	
		G-12K-109A	XRF	483.5	7793.5	
		G-12K-109B	-	-	-	
		G-12K-111A	_	-	_	
		G-12K-112A	-	-	-	
	0/15/00	G-12K-112B	-	-	-	
	8/15/08	G-12K-113A	-	-	-	
		G-12K-114A	_	-	-	
		G-12K-114B	_	-	_	
		G-12K-114C	_	-	_	
		G-12K-115A	XRF	494.5	8288.0	
		G-12K-117A	-	-	-	
		G-12K-117B	_	_	_	
		G-12K-127A	_	_	_	
		G-12K-129A	_	_	_	
		G-12K-138A	XRF, DCP, F	472.5	8760.5	
	8/16/08	G-12K-139A	XRF	112.5	8873.0	

### Table 6.5. List of Glass Discharged, Masses, and Analysis Performed during DM1200 Test (continued).

Table 6.6. XRF Analyzed Composition for Glass Discharged During DM100 Melter Test
(wt%).

Formulation					HI	W-E-A	1-27				
Al Sources					Alumi	num Hy	droxide				
Temperature					120	0 °C				115	0 °C
Test					1	1				2	2
Glass (kg)	Target	28.72	62.52	93.86	117.20	143.26	169.38	196.70	222.18	255.68	277.08
Constituent		BLX-	BLX-	BLX-	BLX-	BLX-	BLX-	BLX-	BLX-	BLX-	BLX-
Constituent		G-67A	G-71A	G-71C	G-76B	G-79A	G-82A	G-84A	G-84C	G-91A	G-96A
Al <sub>2</sub> O <sub>3</sub>	23.97	12.36	13.93	15.26	16.75	17.40	17.96	18.65	19.33	19.77	20.25
B <sub>2</sub> O <sub>3</sub> *	15.19	13.98	14.19	14.35	14.45	14.55	14.64	14.71	14.78	14.85	14.89
BaO	0.05	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01
Bi <sub>2</sub> O <sub>3</sub>	1.14	0.79	0.90	0.95	0.90	0.99	1.09	1.10	1.06	1.05	1.12
CaO	6.08	2.67	3.35	3.74	3.89	4.34	4.69	4.86	4.98	5.05	5.25
CdO	0.02	0.04	0.04	0.04	0.03	0.03	0.03	0.03	0.03	0.03	0.03
$Ce_2O_3$	§	0.04	0.04	0.03	0.02	0.02	0.01	0.02	0.02	0.02	0.02
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.52	0.54	0.53	0.50	0.51	0.56	0.56	0.55	0.56	0.56
F	0.67	0.22#	0.26	0.29	0.32	0.35#	0.36	0.36	0.37	0.38#	0.39
Fe <sub>2</sub> O <sub>3</sub>	5.90	7.24	7.18	6.97	6.32	6.52	6.78	6.63	6.38	6.16	6.27
K <sub>2</sub> O	0.14	0.29	0.28	0.27	0.24	0.24	0.24	0.24	0.23	0.21	0.21
La <sub>2</sub> O <sub>3</sub>	§	0.07	< 0.01	< 0.01	0.04	0.04	0.03	0.02	0.02	< 0.01	< 0.01
Li <sub>2</sub> O*	3.57	2.60	2.76	2.89	2.97	3.05	3.12	3.19	3.24	3.29	3.33
MgO	0.12	0.35	0.30	0.31	0.31	0.29	0.26	0.30	0.30	0.27	0.26
MnO	0.00	0.72	0.61	0.53	0.41	0.39	0.36	0.32	0.28	0.24	0.22
Na <sub>2</sub> O	9.58	11.07	10.63	10.47	11.28	10.51	9.75	10.01	10.18	10.53	9.97
Nd <sub>2</sub> O <sub>3</sub>	§	0.06	0.05	0.04	0.03	0.03	0.03	0.02	0.02	0.02	0.02
NiO	0.40	0.51	0.51	0.49	0.42	0.43	0.45	0.42	0.41	0.37	0.39
P <sub>2</sub> O <sub>5</sub>	1.05	0.74	0.81	0.85	0.91	0.93	0.99	1.00	1.05	1.04	1.06
PbO	0.41	0.27	0.31	0.31	0.30	0.32	0.35	0.35	0.33	0.33	0.36
SiO <sub>2</sub>	30.50	36.56	35.40	34.79	34.50	33.95	33.45	33.00	32.97	32.71	32.54
SO <sub>3</sub>	0.20	0.12	0.12	0.13	0.12	0.13	0.12	0.12	0.11	0.11	0.13
SrO	§	4.04	3.58	3.06	2.35	2.20	2.08	1.76	1.45	1.25	1.12
TiO <sub>2</sub>	0.01	0.05	0.03	0.03	0.04	0.04	0.04	0.04	0.04	0.04	0.04
ZnO	0.08	0.60	0.52	0.46	0.36	0.34	0.34	0.30	0.27	0.23	0.22
ZrO <sub>2</sub>	0.39	4.09	3.67	3.20	2.51	2.40	2.28	1.98	1.65	1.47	1.37
Sum	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00

 § - Not a target constituent
 \* - for XRF-analyzed compositions, values for B<sub>2</sub>O<sub>3</sub> and Li<sub>2</sub>O were calculated based on simple well-stirred tank model and target values.

# - Fluorine was measured by XRF on polished samples, values for other samples calculated by interpolation

#### Table 6.6. XRF Analyzed Composition for Glass Discharged During DM100 Melter Test (wt%) (continued).

Formulation		HLW-E-Al-27												
Al Source				Al	uminum	Hydroxi	de							
Temperature						1150 °C								
Test						2								
Glass (kg)	Target	300.18 323.94 344.56 376.52 400.76 421.76					457.54	490	0.10					
Constituent		BLX-G- 99A	BLX-G- 102B	BLX-G- 108A	BLX-G- 113A	BLX-G- 114B	BLX-G- 115A	BLX-G- 119A	BLX-G- 119B	%Dev.				
Al <sub>2</sub> O <sub>3</sub>	23.97	20.70	20.88	20.97	21.38	21.37	21.79	21.40	21.90	-8.64				
B <sub>2</sub> O <sub>3</sub> *	15.19	14.92	14.96	14.98	15.01	15.04	15.05	15.05	15.05	NC				
BaO	0.05	< 0.01	0.05	< 0.01	< 0.01	0.05	0.05	0.05	0.04	NC				
Bi <sub>2</sub> O <sub>3</sub>	1.14	1.06	1.11	1.13	1.12	1.17	1.13	1.24	1.10	-4.09				
CaO	6.08	5.16	5.36	5.44	5.54	5.72	5.57	5.88	5.55	-8.66				
CdO	0.02	0.03	0.03	0.03	0.02	0.03	0.03	0.03	0.03	NC				
Ce <sub>2</sub> O <sub>3</sub>	§	0.01	0.01	0.02	0.02	0.00	0.02	0.01	< 0.01	NC				
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.53	0.56	0.56	0.56	0.57	0.55	0.59	0.54	NC				
F	0.67	0.40	0.40	0.41#	0.42	0.42	0.43	0.44	0.44#	NC				
Fe <sub>2</sub> O <sub>3</sub>	5.90	5.90	6.00	6.08	6.00	6.14	5.99	6.36	5.75	-2.53				
K <sub>2</sub> O	0.14	0.19	0.20	0.20	0.19	0.19	0.19	0.20	0.19	NC				
La <sub>2</sub> O <sub>3</sub>	§	< 0.01	< 0.01	0.01	0.01	0.01	< 0.01	< 0.01	0.01	NC				
Li <sub>2</sub> O*	3.57	3.35	3.38	3.40	3.43	3.45	3.46	3.46	3.46	NC				
MgO	0.12	0.22	0.25	0.24	0.27	0.23	0.25	0.20	0.26	NC				
MnO	0.00	0.18	0.17	0.16	0.13	0.13	0.12	0.15	0.12	NC				
Na <sub>2</sub> O	9.58	10.46	10.40	10.31	10.38	10.10	10.43	9.47	10.54	10.03				
Nd <sub>2</sub> O <sub>3</sub>	§	0.02	0.01	< 0.01	< 0.01	< 0.01	0.01	0.01	0.01	NC				
NiO	0.40	0.35	0.36	0.36	0.36	0.37	0.35	0.38	0.33	NC				
P <sub>2</sub> O <sub>5</sub>	1.05	1.08	1.10	1.11	1.12	1.15	1.11	1.13	1.16	10.06				
PbO	0.41	0.33	0.35	0.36	0.35	0.37	0.36	0.39	0.34	NC				
SiO <sub>2</sub>	30.50	32.59	32.05	32.06	31.75	31.61	31.42	31.65	31.59	3.60				
SO <sub>3</sub>	0.20	0.14	0.14	0.15	0.15	0.14	0.16	0.15	0.15	NC				
SrO	§	0.93	0.86	0.77	0.65	0.62	0.54	0.61	0.49	NC				
TiO <sub>2</sub>	0.01	0.03	0.04	0.04	0.04	0.04	0.04	0.04	0.04	NC				
ZnO	0.08	0.19	0.18	0.17	0.16	0.16	0.14	0.17	0.14	NC				
ZrO <sub>2</sub>	0.39	1.21	1.14	1.04	0.92	0.92	0.84	0.94	0.77	NC				
Sum	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	NC				

 § - Not a target constituent
 \* - for XRF-analyzed compositions, values for B<sub>2</sub>O<sub>3</sub> and Li<sub>2</sub>O were calculated based on simple well-stirred tank model and target values.

# - Fluorine was measured by XRF on polished samples, values for other samples calculated by interpolation NC – Not calculated

### Table 6.6. XRF Analyzed Composition for Glass Discharged During DM100BL Melter Test (wt%) (continued).

Formulation	HWI-Al-16												
Al Source				Al	uminum		de						
Temperature						1200 °C							
Test						3							
Glass (kg)	Target	510.26	534.52	559.24	588.14	613.16	636.14	659.64	683.14	715.02			
Constituent		BLX-G- 144A	BLX-G- 145A		BLX-G- 151B	BLX-G- 153A	BLY-G- 8A	BLY-G- 10A	BLY-G- 10C	BLY-G- 14B			
Al <sub>2</sub> O <sub>3</sub>	23.25	21.54	21.85	147A 21.79	22.00	21.89	8A 21.81	21.87	22.03	22.12			
$\frac{A_{12}O_3}{B_2O_3*}$	17.73	15.34	15.64	15.91	16.18	16.38	16.54	16.69	16.82	16.97			
BaO	0.05	0.04	0.04	0.05	0.05	0.06	0.05	0.05	0.04				
		1.22	1.17				1.22			0.06			
Bi <sub>2</sub> O <sub>3</sub> CaO	1.11 5.89	5.97	5.85	1.14 5.72	1.13 5.74	1.18 5.84	5.98	1.16 5.91	1.15 5.82	5.77			
CdO	0.02	0.03	0.03	0.03	0.03	0.03	0.03	0.03	0.02	0.03			
Ce <sub>2</sub> O <sub>3</sub>	§	< 0.01	< 0.01	0.05	0.03	< 0.05	0.05	0.05	0.02	< 0.05			
$Cr_2O_3$	8 0.51	0.46	0.46	0.46	0.01	0.48	0.50	0.52	0.51	0.50			
F	0.65	0.36#	0.37	0.38	0.39	0.40 <sup>#</sup>	0.39	0.32	0.37	0.36#			
Fe <sub>2</sub> O <sub>3</sub>	5.72	6.15	5.93	5.80	5.76	5.83	6.09	5.99	5.89	5.75			
K <sub>2</sub> O	0.14	0.13	0.18	0.17	0.18	0.17	0.18	0.18	0.17	0.16			
La <sub>2</sub> O <sub>3</sub>	§	< 0.01	0.02	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	<0.01	0.01			
Li <sub>2</sub> O*	3.46	3.46	3.46	3.46	3.46	3.46	3.46	3.46	3.46	3.46			
MgO	0.11	0.30	0.32	0.29	0.34	0.29	0.27	0.32	0.30	0.30			
MnO	§	0.20	0.18	0.15	0.14	0.13	0.12	0.11	0.10	0.09			
Na <sub>2</sub> O	9.29	9.60	9.71	10.17	9.76	9.57	9.53	9.67	9.45	9.48			
Nd <sub>2</sub> O <sub>3</sub>	ş	0.01	0.01	0.01	< 0.01	< 0.01	0.01	0.01	< 0.01	< 0.01			
NiO	0.39	0.31	0.32	0.32	0.29	0.31	0.34	0.35	0.34	0.32			
P <sub>2</sub> O <sub>5</sub>	1.02	1.11	1.12	1.12	1.15	1.11	1.11	1.13	1.12	1.16			
PbO	0.40	0.40	0.37	0.37	0.36	0.38	0.39	0.37	0.36	0.36			
SiO <sub>2</sub>	29.58	31.60	31.45	31.28	31.31	31.20	30.75	30.71	30.94	30.97			
SO <sub>3</sub>	0.19	0.13	0.12	0.11	0.12	0.12	0.11	0.10	0.10	0.11			
SrO	ş	0.47	0.39	0.32	0.28	0.27	0.24	0.19	0.18	0.15			
TiO <sub>2</sub>	0.01	0.05	0.05	0.04	0.04	0.04	0.04	0.04	0.05	0.04			
ZnO	0.08	0.18	0.16	0.14	0.13	0.13	0.13	0.13	0.12	0.10			
ZrO <sub>2</sub>	0.38	0.86	0.80	0.75	0.69	0.72	0.71	0.63	0.62	0.59			
Sum	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00			

§ - Not a target constituent

\* - for XRF-analyzed compositions, values for B<sub>2</sub>O<sub>3</sub> and Li<sub>2</sub>O were calculated based on simple well-stirred tank model and target values.

# - Fluorine was measured by XRF on polished samples, values for other samples calculated by interpolation

## Table 6.6. XRF Analyzed Composition for Glass Discharged During DM100 Melter Test (wt%) (continued).

Formulation					н	WI-Al-1	6				
Al Source						um Hyo					
Temperature						v	0°C				
Test							<u> </u>				
Glass (kg)	Target	736.80	756.48	777.68	796.34	819.48	-	856.96	875.60	896.38	920.66
	-				BLY-G-						
Constituent		16A	17A	22A	25A	26A	26C	28B	34A	35A	36B
Al <sub>2</sub> O <sub>3</sub>	23.25	22.23	22.03	21.96	21.91	22.26	22.01	22.28	21.99	22.04	22.09
B <sub>2</sub> O <sub>3</sub> *	17.73	17.05	17.12	17.19	17.25	17.30	17.35	17.39	17.42	17.45	17.49
BaO	0.05	0.06	0.05	0.07	0.06	0.06	0.06	0.06	0.06	0.06	0.06
Bi <sub>2</sub> O <sub>3</sub>	1.11	1.08	1.17	1.15	1.21	1.15	1.16	1.12	1.19	1.19	1.22
CaO	5.89	5.64	5.90	5.90	6.07	5.85	5.86	5.76	6.00	5.98	6.14
CdO	0.02	0.03	0.03	0.03	0.03	0.03	0.03	0.03	0.03	0.03	0.03
$Ce_2O_3$	§	0.01	0.01	< 0.01	< 0.01	0.01	0.01	0.01	0.01	0.01	0.01
Cr <sub>2</sub> O <sub>3</sub>	0.51	0.47	0.50	0.52	0.53	0.52	0.53	0.52	0.55	0.55	0.56
F	0.65	0.37	0.38	0.39	$0.40^{\#}$	0.40	0.41	0.41	0.41#	0.42	0.44
Fe <sub>2</sub> O <sub>3</sub>	5.72	5.55	5.84	5.88	6.06	5.83	5.88	5.71	6.04	6.03	6.10
K <sub>2</sub> O	0.14	0.16	0.17	0.17	0.17	0.16	0.16	0.16	0.16	0.17	0.17
$La_2O_3$	§	0.02	< 0.01	< 0.01	< 0.01	< 0.01	0.01	< 0.01	< 0.01	0.01	< 0.01
Li <sub>2</sub> O*	3.46	3.46	3.46	3.46	3.46	3.46	3.46	3.46	3.46	3.46	3.46
MgO	0.11	0.33	0.26	0.32	0.32	0.30	0.32	0.30	0.31	0.33	0.28
MnO	ş	0.08	0.08	0.08	0.08	0.07	0.07	0.06	0.06	0.07	0.06
Na <sub>2</sub> O	9.29	9.88	9.69	9.59	9.07	9.23	9.61	9.77	9.33	9.18	8.71
Nd <sub>2</sub> O <sub>3</sub>	ş	< 0.01	< 0.01	0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	0.01	< 0.01
NiO	0.39	0.30	0.32	0.33	0.34	0.33	0.32	0.33	0.35	0.34	0.33
$P_2O_5$	1.02	1.13	1.15	1.12	1.16	1.15	1.13	1.14	1.12	1.13	1.16
PbO	0.40	0.35	0.37	0.38	0.40	0.37	0.37	0.35	0.39	0.39	0.39
SiO <sub>2</sub>	29.58	30.85	30.52	30.50	30.54	30.60	30.30	30.27	30.19	30.24	30.39
$SO_3$	0.19	0.10	0.11	0.11	0.11	0.12	0.12	0.13	0.14	0.13	0.13
SrO	ş	0.13	0.12	0.11	0.11	0.09	0.10	0.08	0.08	0.07	0.07
TiO <sub>2</sub>	0.01	0.05	0.04	0.05	0.04	0.04	0.04	0.04	0.04	0.05	0.04
ZnO	0.08	0.11	0.10	0.11	0.10	0.10	0.10	0.09	0.10	0.10	0.10
ZrO <sub>2</sub>	0.38	0.55	0.58	0.58	0.60	0.55	0.58	0.53	0.56	0.56	0.56
Sum	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00

§ - Not a target constituent

\* - for XRF-analyzed compositions, values for B<sub>2</sub>O<sub>3</sub> and Li<sub>2</sub>O were calculated based on simple well-stirred tank model and target values.

# - Fluorine was measured by XRF on polished samples, values for other samples calculated by interpolation

#### Table 6.6. XRF Analyzed Composition for Glass Discharged During DM100 Melter Test (wt%) (continued).

Formulation					Н	WI-Al-1	16							
Al Source		Ah	ıminum	Hydrox	ide	Aluminum Oxide								
Temperature			115	) °C			1200 °C							
Test	Torgot		4	1		5								
Glass (kg)	Target	949.08	978.22	995	5.94	1024.64	1057.50	1084.16	1110.42	1139.06	1166.20			
Constituent		BLY- G-40A	BLY- G-40B	BLY- G-40C	%Dev.	BLY- G-63A	BLY- G-66B	BLY- G-69A	BLY- G-73A	BLY- G-75A	BLY- G-80A			
Al <sub>2</sub> O <sub>3</sub>	23.25	22.07	21.96	21.78	-6.33	22.18	22.18	22.18	22.13	22.13	22.58			
B <sub>2</sub> O <sub>3</sub> *	17.73	17.52	17.52	17.52	NC	17.56	17.59	17.61	17.62	17.64	17.65			
BaO	0.05	0.06	0.05	0.06	NC	0.05	0.05	0.05	0.05	0.06	0.06			
Bi <sub>2</sub> O <sub>3</sub>	1.11	1.14	1.17	1.22	9.70	1.11	1.16	1.16	1.19	1.19	1.07			
CaO	5.89	5.93	5.98	6.19	4.97	5.74	5.88	5.90	5.90	5.99	5.56			
CdO	0.02	0.03	0.03	0.03	NC	0.02	0.02	0.02	0.03	0.03	0.02			
Ce <sub>2</sub> O <sub>3</sub>	ş	< 0.01	< 0.01	0.01	NC	< 0.01	0.01	0.01	0.01	< 0.01	< 0.01			
Cr <sub>2</sub> O <sub>3</sub>	0.51	0.53	0.53	0.54	NC	0.44	0.46	0.49	0.50	0.50	0.46			
F	0.65	0.46	0.48#	0.48	NC	0.39#	0.39	0.39	0.39	0.39#	0.40			
Fe <sub>2</sub> O <sub>3</sub>	5.72	5.84	5.94	6.16	7.62	6.00	6.25	6.29	6.41	6.54	5.99			
K <sub>2</sub> O	0.14	0.17	0.17	0.18	NC	0.17	0.17	0.17	0.17	0.17	0.16			
La <sub>2</sub> O <sub>3</sub>	ş	< 0.01	< 0.01	< 0.01	NC	< 0.01	< 0.01	< 0.01	< 0.01	0.01	0.01			
Li <sub>2</sub> O*	3.46	3.46	3.46	3.46	NC	3.46	3.46	3.46	3.46	3.46	3.46			
MgO	0.11	0.38	0.35	0.30	NC	0.29	0.28	0.26	0.25	0.21	0.26			
MnO	§	0.06	0.07	0.08	NC	0.09	0.08	0.07	0.07	0.07	0.06			
Na <sub>2</sub> O	9.29	9.56	9.37	8.66	-6.79	9.89	9.12	9.32	9.40	9.03	9.56			
$Nd_2O_3$	ş	0.01	< 0.01	< 0.01	NC	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01			
NiO	0.39	0.33	0.34	0.34	NC	0.30	0.30	0.32	0.32	0.32	0.29			
$P_2O_5$	1.02	1.13	1.12	1.15	12.72	1.10	1.13	1.11	1.10	1.13	1.14			
PbO	0.40	0.38	0.38	0.41	NC	0.35	0.35	0.35	0.36	0.35	0.32			
SiO <sub>2</sub>	29.58	30.09	30.19	30.44	2.92	29.89	30.11	29.82	29.58	29.72	29.96			
$SO_3$	0.19	0.13	0.13	0.13	NC	0.11	0.11	0.11	0.10	0.11	0.11			
SrO	ş	0.06	0.07	0.09	NC	0.07	0.06	0.05	0.05	0.04	0.03			
TiO <sub>2</sub>	0.01	0.04	0.05	0.05	NC	0.04	0.04	0.05	0.04	0.05	0.04			
ZnO	0.08	0.09	0.10	0.11	NC	0.10	0.10	0.10	0.10	0.10	0.08			
ZrO <sub>2</sub>	0.38	0.53	0.55	0.61	NC	0.65	0.69	0.71	0.76	0.75	0.70			
Sum	100.00	100.00	100.00	100.00	NC	100.00	100.00	100.00	100.00	100.00	100.00			

 § - Not a target constituent
 \* - for XRF-analyzed compositions, values for B<sub>2</sub>O<sub>3</sub> and Li<sub>2</sub>O were calculated based on simple well-stirred tank model and target values.

# - Fluorine was measured by XRF on polished samples, values for other samples calculated by interpolation NC - Not calculated

#### Table 6.6. XRF Analyzed Composition for Glass Discharged During DM100 Melter Test (wt%) (continued).

Formulation					HWI-	Al-16				
Al Source					Aluminu	m Oxide	;			
Temperature		1200 °C				115	0 °C			
Test		5					6			
Glass (kg)	Target	1193.60	1227.24	1249.92	1270.34	1286.12	1306.46	1322.02	1350.64	1374.18
Constituent		BLY-G- 80C	BLY-G- 85A	BLY-G- 91A	BLY-G- 93A	BLY-G- 93C	BLY-G- 98B	BLY-G- 101A	BLY-G- 107A	BLY-G- 111B
Al <sub>2</sub> O <sub>3</sub>	23.25	22.38	22.25	22.37	22.36	22.53	22.43	22.49	22.67	22.23
$\frac{\text{Al}_2\text{O}_3}{\text{B}_2\text{O}_3*}$	17.73	17.66	17.68	17.68	17.69	17.69	17.70	17.70	17.71	17.71
BaO	0.05	0.05	0.05	0.07	0.06	0.06	0.05	0.05	0.05	0.05
BaO Bi <sub>2</sub> O <sub>3</sub>	1.11	1.16	1.18	1.16	1.13	1.10	1.12	1.11	1.08	1.22
CaO	5.89	5.91	5.83	5.79	5.70	5.69	5.76	5.73	5.63	5.98
CdO	0.02	0.03	0.03	0.03	0.03	0.03	0.02	0.03	0.02	0.02
Ce <sub>2</sub> O <sub>3</sub>	§	< 0.05	0.05	0.05	< 0.05	0.05	<0.02	0.03	<0.02	0.02
$Cr_2O_3$	8 0.51	0.50	0.50	0.49	0.50	0.50	0.50	0.48	0.49	0.54
F	0.65	0.30	0.43	0.44#	0.30	0.30	0.30	0.46	0.47#	0.47
Fe <sub>2</sub> O <sub>3</sub>	5.72	6.44	6.47	6.37	6.30	6.22	6.31	6.24	6.12	6.82
K <sub>2</sub> O	0.14	0.17	0.47	0.17	0.16	0.22	0.17	0.24	0.12	0.02
La <sub>2</sub> O <sub>3</sub>	§	<0.01	< 0.01	<0.01	< 0.01	<0.01	<0.01	<0.01	<0.01	< 0.01
Li <sub>2</sub> O*	3.46	3.46	3.46	3.46	3.46	3.46	3.46	3.46	3.46	3.46
MgO	0.11	0.23	0.23	0.22	0.26	0.21	0.27	0.20	0.23	0.22
MnO	§	0.06	0.05	0.05	0.05	0.05	0.05	0.04	0.04	0.04
Na <sub>2</sub> O	9.29	8.93	9.30	9.28	9.58	9.33	9.20	9.26	9.34	9.13
Nd <sub>2</sub> O <sub>3</sub>	ş	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01
NiO	0.39	0.33	0.32	0.30	0.29	0.30	0.30	0.30	0.28	0.34
P <sub>2</sub> O <sub>5</sub>	1.02	1.15	1.20	1.17	1.16	1.18	1.17	1.18	1.18	1.12
PbO	0.40	0.34	0.34	0.35	0.34	0.32	0.32	0.33	0.31	0.36
SiO <sub>2</sub>	29.58	29.75	29.47	29.55	29.48	29.69	29.71	29.73	29.78	29.00
SO <sub>3</sub>	0.19	0.11	0.10	0.11	0.10	0.11	0.12	0.12	0.13	0.12
SrO	§	0.03	0.03	0.03	0.02	0.02	0.02	0.02	0.02	0.02
TiO <sub>2</sub>	0.01	0.05	0.05	0.04	0.04	0.04	0.04	0.04	0.05	0.05
ZnO	0.08	0.09	0.09	0.08	0.08	0.08	0.08	0.08	0.08	0.08
ZrO <sub>2</sub>	0.38	0.75	0.76	0.77	0.76	0.75	0.75	0.76	0.71	0.83
Sum	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00

 § - Not a target constituent
 \* - for XRF-analyzed compositions, values for B<sub>2</sub>O<sub>3</sub> and Li<sub>2</sub>O were calculated based on simple well-stirred tank model and target values.

# - Fluorine was measured by XRF on polished samples, values for other samples calculated by interpolation

## Table 6.6. XRF Analyzed Composition for Glass Discharged During DM100 Melter Test (wt%) (continued)

Formulation		H	WI-Al-1	6			H	WI-Al-1	9	
Al Source		Alur	ninum O	xide			Alumi	num Hyd	lroxide	
Temperature			115	0 °C				120	0 °C	
Test			(	6					7	
Glass (kg)	Target	1395.94	1423.10	1464	4.14	Target	1496.38	1529.96	1560.12	1591.50
Constituent		BLY-G- 114A	BLY-G- 120A	BLY-G- 120B	%Dev.		BLY-G- 139A	BLY-G- 141B	BLY-G- 144B	BLY-G- 145B
Al <sub>2</sub> O <sub>3</sub>	23.25	22.36	22.38	22.45	-3.46	23.97	22.31	22.80	22.66	22.67
B <sub>2</sub> O <sub>3</sub> *	17.73	17.71	17.71	17.71	NC	19.19	17.95	18.16	18.32	18.46
BaO	0.05	0.05	0.05	0.05	NC	0.05	0.06	0.06	0.06	0.07
Bi <sub>2</sub> O <sub>3</sub>	1.11	1.18	1.16	1.15	3.38	1.14	1.29	1.21	1.26	1.29
CaO	5.89	5.84	5.81	5.74	-2.63	5.58	6.11	5.85	6.08	6.06
CdO	0.02	0.03	0.02	0.02	NC	0.02	0.02	0.03	0.02	0.03
Ce <sub>2</sub> O <sub>3</sub>	ş	< 0.01	< 0.01	0.02	NC	ş	0.02	< 0.01	< 0.01	0.01
Cr <sub>2</sub> O <sub>3</sub>	0.51	0.52	0.51	0.49	NC	0.52	0.59	0.57	0.59	0.59
F	0.65	0.47	0.47	0.47	NC	0.67	0.35#	0.37	0.38	0.39
Fe <sub>2</sub> O <sub>3</sub>	5.72	6.55	6.32	6.27	9.58	5.90	6.76	6.33	6.60	6.52
K <sub>2</sub> O	0.14	0.17	0.17	0.17	NC	0.14	0.18	0.18	0.19	0.19
La <sub>2</sub> O <sub>3</sub>	ş	< 0.01	< 0.01	< 0.01	NC	§	< 0.01	< 0.01	< 0.01	< 0.01
Li <sub>2</sub> O*	3.46	3.46	3.46	3.46	NC	3.57	3.48	3.50	3.51	3.52
MgO	0.11	0.21	0.21	0.19	NC	0.12	0.29	0.31	0.28	0.28
MnO	ş	0.04	0.05	0.07	NC	ş	0.05	0.04	0.04	0.04
Na <sub>2</sub> O	9.29	9.04	9.53	9.42	1.41	9.58	9.38	9.44	8.89	9.05
Nd <sub>2</sub> O <sub>3</sub>	ş	< 0.01	< 0.01	< 0.01	NC	§	0.01	< 0.01	< 0.01	< 0.01
NiO	0.39	0.31	0.30	0.29	NC	0.40	0.38	0.34	0.36	0.35
$P_2O_5$	1.02	1.15	1.13	1.13	10.58	1.05	1.03	1.02	1.01	0.97
PbO	0.40	0.34	0.33	0.33	NC	0.41	0.38	0.35	0.37	0.39
SiO <sub>2</sub>	29.58	29.51	29.36	29.52	-0.21	27.00	28.35	28.52	28.44	28.17
SO <sub>3</sub>	0.19	0.12	0.10	0.11	NC	0.20	0.10	0.10	0.10	0.10
SrO	ş	0.02	0.02	0.03	NC	ş	0.02	0.02	0.02	0.02
TiO <sub>2</sub>	0.01	0.04	0.04	0.05	NC	0.01	0.05	0.05	0.05	0.05
ZnO	0.08	0.08	0.08	0.09	NC	0.08	0.10	0.09	0.09	0.10
ZrO <sub>2</sub>	0.38	0.80	0.77	0.78	NC	0.39	0.74	0.69	0.69	0.69
Sum	100.00	100.00	100.00	100.00	NC	100.00	100.00	100.00	100.00	100.00

§ - Not a target constituent

\* - for XRF-analyzed compositions, values for B<sub>2</sub>O<sub>3</sub> and Li<sub>2</sub>O were calculated based on simple well-stirred tank model and target values.

# - Fluorine was measured by XRF on polished samples, values for other samples calculated by interpolation NC – Not calculated

## Table 6.6. XRF Analyzed Composition for Glass Discharged During DM100 Melter Test (wt%) (continued).

Formulation		HWI-Al-19											
Al Source				Al	uminum		de						
Temperature			120					1150 °C					
Test			7					8					
Glass (kg)	Target	1625.48	1655.82	1681.98	1709.34	1741.30	1767.04	1791.66	1815.98	1841.26			
Constituent		BLY-G- 151A	BLY-G- 152A	BLY-G- 153B	BLZ-G- 5B	BLZ-G- 13A	BLZ-G- 16A	BLZ-G- 21A	BLZ-G- 23A	BLZ-G- 27A			
Al <sub>2</sub> O <sub>3</sub>	23.97	22.53	22.52	22.82	22.73	22.77	22.65	21A 22.72	23A 22.85	27A 22.82			
$\frac{\text{Al}_2\text{O}_3}{\text{B}_2\text{O}_3*}$	19.19	18.59	18.68	18.75	18.81	18.87	18.92	18.95	18.98	19.01			
B <sub>2</sub> O <sub>3</sub> . BaO	0.05	0.06	0.06	0.05	0.05	0.07	0.07	0.06	0.06	0.06			
BaO Bi <sub>2</sub> O <sub>3</sub>	1.14	1.28	1.23	1.21	1.25	1.26	1.31	1.27	1.26	1.27			
CaO	5.58	6.00	5.93	5.77	5.86	5.88	5.98	5.91	5.85	5.81			
CdO	0.02	0.00	0.03	0.02	0.02	0.03	0.03	0.03	0.03	0.03			
$Ce_2O_3$	<u> </u>	< 0.03	< 0.03	0.02	0.02	0.03	< 0.03	0.03	0.03	< 0.03			
$Cr_2O_3$	0.52	0.61	0.58	0.57	0.60	0.58	0.59	0.59	0.58	0.58			
F	0.52	0.41#	0.38	0.37	0.40	0.40#	0.39	0.39	0.38	0.38			
Fe <sub>2</sub> O <sub>3</sub>	5.90	6.50	6.32	5.97	6.22	6.26	6.38	6.17	6.18	6.07			
K <sub>2</sub> O	0.14	0.19	0.18	0.18	0.22	0.20	0.19	0.17	0.18	0.07			
$La_2O_3$	§	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01			
Li <sub>2</sub> O*	3.57	3.53	3.53	3.54	3.54	3.55	3.55	3.55	3.56	3.56			
MgO	0.12	0.28	0.34	0.35	0.33	0.32	0.28	0.30	0.35	0.28			
MnO	§	0.03	0.04	0.03	0.03	0.03	0.03	0.03	0.03	0.03			
Na <sub>2</sub> O	<u> </u>	9.40	9.50	9.41	9.78	9.37	9.29	9.43	9.56	9.59			
Nd <sub>2</sub> O <sub>3</sub>	\$.50	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01			
NiO	0.40	0.37	0.37	0.32	0.35	0.35	0.36	0.34	0.36	0.34			
P <sub>2</sub> O <sub>5</sub>	1.05	0.96	0.96	1.02	0.97	0.98	0.99	0.97	0.96	0.97			
PbO	0.41	0.37	0.38	0.36	0.38	0.39	0.39	0.39	0.37	0.37			
SiO <sub>2</sub>	27.00	27.96	28.06	28.40	27.65	27.83	27.71	27.80	27.56	27.73			
SO <sub>3</sub>	0.20	0.11	0.10	0.11	0.11	0.11	0.13	0.13	0.13	0.13			
SrO	§	0.02	0.02	0.01	0.01	0.01	0.01	0.01	0.01	0.01			
TiO <sub>2</sub>	0.01	0.05	0.05	0.04	0.05	0.05	0.05	0.04	0.05	0.04			
ZnO	0.08	0.09	0.09	0.09	0.09	0.09	0.09	0.08	0.09	0.08			
ZrO <sub>2</sub>	0.39	0.64	0.64	0.57	0.58	0.61	0.61	0.60	0.58	0.59			
Sum	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00			

§ - Not a target constituent

\* - for XRF-analyzed compositions, values for B<sub>2</sub>O<sub>3</sub> and Li<sub>2</sub>O were calculated based on simple well-stirred tank model and target values.

# - Fluorine was measured by XRF on polished samples, values for other samples calculated by interpolation

Formulation				HWI-Al-19			
Al Source			Alur	ninum Hydro	oxide		
Temperature			11111	v	0°C		
Test					8		
Glass (kg)	Target	1866.96	1889.56	1912.62	1936.76	1948	8 40
Constituent				BLZ-G-33B			% Dev.
Al <sub>2</sub> O <sub>3</sub>	23.97	22.88	23.11	23.03	23.01	22.94	-4.30
B <sub>2</sub> O <sub>3</sub> *	19.19	19.03	19.05	19.07	19.08	19.09	NC
BaO	0.05	0.06	0.05	0.06	0.06	0.05	NC
Bi <sub>2</sub> O <sub>3</sub>	1.14	1.25	1.20	1.28	1.27	1.26	9.96
CaO	5.58	5.87	5.64	5.80	5.80	5.74	2.89
CdO	0.02	0.02	0.02	0.03	0.03	0.02	NC
Ce <sub>2</sub> O <sub>3</sub>	ş	< 0.01	0.01	< 0.01	< 0.01	< 0.01	NC
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.60	0.56	0.59	0.60	0.60	NC
F	0.67	0.45	0.46	0.47	$0.48^{\#}$	0.48	NC
Fe <sub>2</sub> O <sub>3</sub>	5.90	6.17	5.74	6.18	6.11	6.15	4.21
K <sub>2</sub> O	0.14	0.19	0.19	0.19	0.19	0.18	NC
La <sub>2</sub> O <sub>3</sub>	Ş	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	NC
Li <sub>2</sub> O*	3.57	3.56	3.56	3.56	3.56	3.56	NC
MgO	0.12	0.31	0.31	0.32	0.29	0.31	NC
MnO	ş	0.03	0.03	0.03	0.03	0.03	NC
Na <sub>2</sub> O	9.58	9.25	9.88	9.23	8.81	9.54	-0.38
Nd <sub>2</sub> O <sub>3</sub>	Ş	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	NC
NiO	0.40	0.36	0.31	0.35	0.34	0.35	NC
P <sub>2</sub> O <sub>5</sub>	1.05	0.97	0.99	0.95	0.99	0.94	-10.22
PbO	0.41	0.38	0.35	0.37	0.37	0.36	NC
SiO <sub>2</sub>	27.00	27.77	27.74	27.65	28.14	27.55	2.04
SO <sub>3</sub>	0.20	0.13	0.13	0.14	0.14	0.14	NC
SrO	ş	0.01	0.01	0.01	0.01	0.01	NC
TiO <sub>2</sub>	0.01	0.05	0.04	0.04	0.04	0.04	NC
ZnO	0.08	0.08	0.08	0.08	0.08	0.08	NC
ZrO <sub>2</sub>	0.39	0.57	0.52	0.56	0.57	0.57	NC
Sum	100.00	100.00	100.00	100.00	100.00	100.00	NC

# Table 6.6. XRF Analyzed Composition for Glass Discharged During DM100 Melter Test (wt%) (continued).

§ - Not a target constituent

\* - for XRF-analyzed compositions, values for B<sub>2</sub>O<sub>3</sub> and Li<sub>2</sub>O were calculated based on simple well-stirred tank model and target values.

# - Fluorine was measured by XRF on polished samples, values for other samples calculated by interpolation NC – Not calculated

Test					1					2	
Glass (kg)	Target	478.5	980.5		1961.0	2446.0		3373.0	3865.5	4364.0	
Constituent	0.1	G-12J-	G-12J-	G-12J-	G-12J-	G-12J-	G-12J-	G-12J-	G-12J-		G-12K-
		45A	60A	64A	72A	79B	100A	105B	135A	149B	21A
$Al_2O_3$	23.97	13.80	16.12	17.29	18.51	19.35	20.23	20.82	21.07	21.71	22.08
$B_2O_3*$	19.19	13.58	15.81	16.56	17.16	17.61	17.94	18.21	18.43	18.61	18.74
BaO	0.05	0.04	0.05	0.04	0.04	0.04	0.04	0.05	0.06	0.05	0.06
Bi <sub>2</sub> O <sub>3</sub>	1.14	0.44	0.61	0.70	0.80	0.94	0.96	1.06	1.09	1.10	1.12
CaO	5.58	2.44	3.18	3.60	3.97	4.44	4.66	4.92	5.16	5.12	5.24
CdO	0.02	0.06	0.05	0.05	0.04	0.04	0.03	0.04	0.04	0.03	0.03
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.32	0.39	0.41	0.43	0.48	0.48	0.51	0.53	0.51	0.54
F	0.67	0.14 <sup>#</sup>	0.19	0.23	0.28#	0.31	0.31	0.30#	0.32	0.33	0.35
Fe <sub>2</sub> O <sub>3</sub>	5.90	8.94	8.21	7.78	7.26	7.24	6.96	6.87	6.93	6.49	6.43
K <sub>2</sub> O	0.14	0.24	0.22	0.23	0.21	0.22	0.21	0.20	0.20	0.20	0.19
$La_2O_3$	§	0.18	0.14	0.13	0.09	0.08	0.07	0.05	0.04	0.03	0.03
Li <sub>2</sub> O*	3.57	3.46	2.90	3.05	3.17	3.26	3.32	3.38	3.42	3.45	3.48
MgO	0.12	0.48	0.40	0.37	0.41	0.38	0.33	0.33	0.34	0.36	0.30
MnO	0.00	0.72	0.56	0.47	0.36	0.31	0.25	0.20	0.17	0.14	0.11
Na <sub>2</sub> O	9.58	11.03	10.18	10.04	10.37	10.00	9.94	9.84	9.48	9.62	9.74
$Nd_2O_3$	§	0.12	0.09	0.07	0.06	0.05	0.04	0.03	0.03	0.02	0.01
NiO	0.40	0.32	0.33	0.33	0.32	0.32	0.32	0.34	0.35	0.31	0.33
P <sub>2</sub> O5	1.05	0.52	0.69	0.75	0.86	0.90	0.94	1.00	1.07	1.07	1.12
PbO	0.41	0.16	0.20	0.22	0.25	0.28	0.29	0.32	0.32	0.31	0.33
$Sb_2O_3$	§	0.04	0.04	0.02	0.02	0.02	0.01	0.01	0.01	0.01	< 0.01
SiO <sub>2</sub>	27.00	39.68	36.98	35.31	33.38	31.92	31.09	30.10	29.61	29.36	28.65
SO3	0.20	0.08	0.10	0.11	0.11	0.12	0.13	0.14	0.12	0.12	0.13
SrO	ş	0.21	0.16	0.14	0.11	0.09	0.07	0.06	0.05	0.04	0.03
TiO <sub>2</sub>	0.01	0.15	0.12	0.11	0.10	0.09	0.08	0.07	0.07	0.07	0.06
ZnO	0.08	1.10	0.86	0.72	0.57	0.49	0.41	0.34	0.30	0.25	0.21
ZrO <sub>2</sub>	0.39	1.74	1.42	1.27	1.11	1.03	0.89	0.83	0.78	0.70	0.68
Sum	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00

#### Table 6.7. XRF Analyzed Composition for Glass Discharged During DM1200 Melter Test (wt%).

§ - Not a target constituent
\* - Target values calculated by simple well-stirred tank model using the DCP analyzed values for the first discharge of the whole test.

# - Fluorine was measured by XRF on polished samples, values for other samples calculated by interpolation

## Table 6.7. XRF Analyzed Composition for Glass Discharged During DM1200 Melter Test (wt%) (continued).

Test		2				3	3				5022	0072
Glass (kg)	Toract	5357.5	5832.5	6295.0	6808.0	7310.0	7793.5	8288.0	8760.5	8873.0	5832-	-88/3
	Taiget	G-12K-	A	%Dev.								
Constituent		28C	44A	59A	81B	91A	109A	115A	138A	139A	Avg.	70Dev.
$Al_2O_3$	23.97	22.45	22.80	23.00	22.99	23.43	23.67	23.83	23.71	23.64	23.38	-2.51
$B_2O_3$	19.19	18.84	18.92	18.97	19.03	19.06	19.09	19.11	19.13	19.13	19.06	NC
BaO	0.05	0.06	0.06	0.07	0.06	0.06	0.05	0.07	0.06	0.07	0.06	NC
Bi <sub>2</sub> O <sub>3</sub>	1.14	1.12	1.07	1.12	1.19	1.13	1.14	1.17	1.16	1.15	1.14	-0.22
CaO	5.58	5.31	5.16	5.33	5.55	5.52	5.50	5.47	5.44	5.45	5.43	-2.74
CdO	0.02	0.03	0.03	0.03	0.03	0.03	0.02	0.03	0.03	0.03	0.03	NC
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.53	0.52	0.51	0.55	0.54	0.55	0.54	0.51	0.53	0.53	NC
F	0.67	0.37#	0.36	0.35	0.34	0.33#	0.34	0.36	0.37#	0.37	0.35	NC
Fe <sub>2</sub> O <sub>3</sub>	5.90	6.17	5.84	5.92	6.20	6.07	6.08	5.99	6.00	6.07	6.02	2.02
K <sub>2</sub> O	0.14	0.19	0.19	0.19	0.20	0.19	0.19	0.18	0.18	0.19	0.19	NC
$La_2O_3$	Ş	0.02	0.02	0.02	0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	0.01	NC
Li <sub>2</sub> O	3.57	3.50	3.52	3.53	3.54	3.55	3.55	3.56	3.56	3.56	3.54	NC
MgO	0.12	0.30	0.31	0.29	0.28	0.33	0.30	0.28	0.33	0.31	0.30	NC
MnO	0.00	0.09	0.07	0.07	0.06	0.05	0.05	0.04	0.04	0.04	0.05	NC
Na <sub>2</sub> O	9.58	9.76	10.39	9.91	9.40	9.31	9.54	9.50	9.87	9.84	9.72	1.44
$Nd_2O_3$	Ş	0.01	< 0.01	0.01	0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	NC
NiO	0.40	0.31	0.30	0.31	0.34	0.33	0.33	0.31	0.32	0.33	0.32	NC
P <sub>2</sub> O5	1.05	1.11	1.13	1.13	1.14	1.13	1.12	1.14	1.15	1.16	1.14	7.59
PbO	0.41	0.33	0.31	0.32	0.34	0.32	0.31	0.33	0.33	0.34	0.32	NC
Sb <sub>2</sub> O <sub>3</sub>	§	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	< 0.01	NC
SiO <sub>2</sub>	27.00	28.48	28.10	28.04	27.85	27.74	27.37	27.28	27.01	26.99	27.55	2.00
SO3	0.20	0.14	0.13	0.15	0.14	0.16	0.14	0.13	0.13	0.14	0.14	NC
SrO	§	0.02	0.02	0.02	0.01	0.01	< 0.01	< 0.01	0.01	< 0.01	0.01	NC
TiO <sub>2</sub>	0.01	0.06	0.05	0.05	0.05	0.05	0.04	0.05	0.05	0.04	0.05	NC
ZnO	0.08	0.18	0.15	0.14	0.14	0.12	0.11	0.11	0.10	0.11	0.12	NC
ZrO <sub>2</sub>	0.39	0.62	0.55	0.56	0.56	0.53	0.49	0.52	0.50	0.50	0.53	NC
Sum	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	100.00	NC
Jot a target o		-										

§ - Not a target constituent

\* - Target values calculated by simple well-stirred tank model using the DCP analyzed values for the first discharge of the whole test.

# - Fluorine was measured by XRF on polished samples, values for other samples calculated by interpolation NC – Not calculated

## Table 6.8. XRF and DCP Analyzed Compositions of Selected Glass Samples Discharged during DM100 and DM1200 Melter Tests (wt%).

Melter Type					DM1	00BL				
Formulation		HI	W-E-Al	-27			H	WI-Al -1	16	
Test		1	l	2	2			3	4	L I
Constituent	Target	BLX-0	G-84C	BLX-C	G-115A	Target	BLY-	G-14B	BLY-	G-36B
Constituent		XRF	DCP	XRF	DCP		XRF	DCP	XRF	DCP
Al <sub>2</sub> O <sub>3</sub>	23.97	19.33	17.06	21.79	18.76	23.25	22.12	19.65	22.09	19.52
B <sub>2</sub> O <sub>3</sub>	15.19	14.78*	13.70	15.05*	15.29	17.73	16.97*	17.16	17.49*	17.66
BaO	0.05	< 0.01	0.06	0.05	0.06	0.05	0.06	0.06	0.06	0.06
Bi <sub>2</sub> O <sub>3</sub>	1.14	1.06	1.02	1.13	1.11	1.11	1.13	1.11	1.22	1.08
CaO	6.08	4.98	4.50	5.57	5.07	5.89	5.77	5.27	6.14	5.35
CdO	0.02	0.03	NA	0.03	NA	0.02	0.03	NA	0.03	NA
$Ce_2O_3$	§	0.02	NA	0.02	NA	§	< 0.01	NA	0.01	NA
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.55	0.22	0.55	0.22	0.51	0.50	0.21	0.56	0.19
F	0.67	0.37 <sup>\$</sup>	NA	0.43 <sup>\$</sup>	NA	0.65	0.36#	NA	0.44 <sup>\$</sup>	NA
Fe <sub>2</sub> O <sub>3</sub>	5.90	6.38	5.90	5.99	5.38	5.72	5.75	5.33	6.10	5.21
K <sub>2</sub> O	0.14	0.23	0.26	0.19	0.23	0.14	0.16	0.20	0.17	0.19
$La_2O_3$	§	0.02	NA	< 0.01	NA	§	0.01	NA	< 0.01	NA
Li <sub>2</sub> O	3.57	3.24*	3.47	3.46*	3.52	3.46	3.46*	3.19	3.46*	3.19
MgO	0.12	0.30	0.34	0.25	0.31	0.11	0.30	0.38	0.28	0.38
MnO	§	0.28	0.31	0.12	0.13	§	0.09	0.10	0.06	0.06
Na <sub>2</sub> O	9.58	10.18	8.81	10.43	8.71	9.29	9.48	8.40	8.71	8.24
$Nd_2O_3$	§	0.02	NA	0.01	NA	§	< 0.01	NA	< 0.01	NA
NiO	0.40	0.41	0.32	0.35	0.26	0.39	0.32	0.27	0.33	0.24
$P_2O5$	1.05	1.05	0.92	1.11	0.97	1.02	1.16	0.99	1.16	1.03
PbO	0.41	0.33	0.36	0.36	0.39	0.40	0.36	0.39	0.39	0.38
SiO <sub>2</sub>	30.50	32.97	31.92	31.42	29.87	29.58	30.97	29.96	30.39	29.32
$SO_3$	0.20	0.11	NA	0.16	NA	0.19	0.11	NA	0.13	NA
SrO	§	1.45	1.62	0.54	0.65	§	0.15	0.19	0.07	0.08
TiO <sub>2</sub>	0.01	0.04	0.05	0.04	0.05	0.01	0.04	0.05	0.04	0.05
ZnO	0.08	0.27	0.23	0.14	0.13	0.08	0.10	0.10	0.10	0.08
ZrO <sub>2</sub>	0.39	1.65	1.91	0.84	0.99	0.38	0.59	0.71	0.56	0.65
Sum	100.00	100.00	92.98	100.00	92.10	100.00	100.00	93.72	100.00	92.96

§ - Not a target constituent;

NA - Not analyzed

\* - for XRF-analyzed compositions, values for B<sub>2</sub>O<sub>3</sub> and Li<sub>2</sub>O were calculated based on simple well-stirred tank model with target values

# - Fluorine measured by XRF on polished samples

\$ - Fluorine values calculated by interpolation

Meter Type					DM1	00BL				
Formulation		H	WI-Al -1	16			H	IWI-Al-1	9	
Test		5		(	5			7	8	3
Constituent	Target	BLY-0	G-80C	BLY-C	6-114A	Target	BLZ-	G-5B	BLZ-0	G-37A
Constituent		XRF	DCP	XRF	DCP		XRF	DCP	XRF	DCP
$Al_2O_3$	23.25	22.38	20.17	22.36	19.85	23.97	22.73	20.09	22.94	19.89
$B_2O_3$	17.73	17.66*	17.35	17.71*	17.31	19.19	18.81*	19.13	19.09*	19.17
BaO	0.05	0.05	0.06	0.05	0.05	0.05	0.05	0.06	0.05	0.06
Bi <sub>2</sub> O <sub>3</sub>	1.11	1.16	1.10	1.18	1.09	1.14	1.25	1.12	1.26	1.16
CaO	5.89	5.91	5.27	5.84	5.15	5.58	5.86	5.32	5.74	5.10
CdO	0.02	0.03	NA	0.03	NA	0.02	0.02	< 0.01	0.02	< 0.01
$Ce_2O_3$	§	< 0.01	NA	< 0.01	NA	§	0.01	NA	< 0.01	NA
Cr <sub>2</sub> O <sub>3</sub>	0.51	0.50	0.23	0.52	0.22	0.52	0.60	0.24	0.60	0.24
F	0.65	0.41 <sup>\$</sup>	NA	0.47 <sup>\$</sup>	NA	0.67	0.40 <sup>\$</sup>	NA	0.48 <sup>\$</sup>	NA
Fe <sub>2</sub> O <sub>3</sub>	5.72	6.44	5.90	6.55	5.72	5.90	6.22	5.50	6.15	5.32
K <sub>2</sub> O	0.14	0.17	0.20	0.17	0.20	0.14	0.18	0.23	0.18	0.22
$La_2O_3$	§	< 0.01	NA	< 0.01	NA	§	< 0.01	NA	< 0.01	NA
Li <sub>2</sub> O	3.46	3.46*	3.57	3.46*	3.42	3.57	3.54*	3.68	3.56*	3.57
MgO	0.11	0.23	0.29	0.21	0.29	0.12	0.33	0.45	0.31	0.41
MnO	§	0.06	0.06	0.04	0.05	§	0.03	0.03	0.03	0.03
Na <sub>2</sub> O	9.29	8.93	8.19	9.04	8.01	9.58	9.78	9.01	9.54	8.90
Nd <sub>2</sub> O <sub>3</sub>	§	< 0.01	NA	< 0.01	NA	§	< 0.01	NA	< 0.01	NA
NiO	0.39	0.33	0.27	0.31	0.23	0.40	0.35	0.26	0.35	0.25
P <sub>2</sub> O5	1.02	1.15	1.05	1.15	0.93	1.05	0.97	0.81	0.94	0.88
PbO	0.40	0.34	0.35	0.34	0.33	0.41	0.38	0.47	0.36	0.46
SiO <sub>2</sub>	29.58	29.75	28.66	29.51	28.08	27.00	27.65	28.81	27.55	28.11
$SO_3$	0.19	0.11	NA	0.12	NA	0.20	0.11	NA	0.14	NA
SrO	§	0.03	0.04	0.02	0.02	§	0.01	0.01	0.01	< 0.01
TiO <sub>2</sub>	0.01	0.05	0.05	0.04	0.05	0.01	0.05	0.04	0.04	0.04
ZnO	0.08	0.09	0.08	0.08	0.07	0.08	0.09	0.06	0.08	0.06
ZrO <sub>2</sub>	0.38	0.75	0.90	0.80	0.92	0.39	0.58	0.57	0.57	0.57
Sum	100.00	100.00	93.79	100.00	91.99	100.00	100.00	95.89	100.00	94.44

### Table 6.8. XRF and DCP Analyzed Compositions of Selected Glass Samples Dischargedduring DM100 and DM1200 Melter Tests (wt%) (continued).

§ - Not a target constituent;

NA - Not analyzed

\* - for XRF-analyzed compositions, values for  $B_2O_3$  and  $Li_2O$  were calculated based on simple well-stirred tank model with target values

# - Fluorine measured by XRF on polished samples

\$ - Fluorine values calculated by interpolation

Table 6.8. XRF and DCP Analyzed Compositions of Selected Glass Samples Discharged
during DM100 and DM1200 Melter Tests (wt%) (continued).

Melter Type		DM1200												
Formulation				Н	WI-Al-1	9								
Test				1		2	2	3	3					
Constituent	Target	G-12.	J-45A	G-12J	-105B	G-12J	-149B	G-12K	-138A					
Constituent		XRF	DCP	XRF	DCP	XRF	DCP	XRF	DCP					
Al <sub>2</sub> O <sub>3</sub>	23.97	13.81	12.46	20.83	18.47	21.71	18.8	23.71	21.44					
$B_2O_3$	19.19	13.58**	13.58	18.21**	17.75	18.61**	18.07	19.13**	18.86					
BaO	0.05	0.04	0.06	0.05	0.06	0.05	0.06	0.06	0.06					
Bi <sub>2</sub> O <sub>3</sub>	1.14	0.44	0.47	1.06	0.99	1.10	1.06	1.16	1.17					
CaO	5.58	2.44	2.54	4.92	4.36	5.12	4.60	5.44	4.95					
CdO	0.02	0.06	< 0.01	0.04	< 0.01	0.03	NA	0.03	< 0.01					
Ce <sub>2</sub> O <sub>3</sub>	§	0.01	NA	< 0.01	NA	0.01	NA	< 0.01	NA					
Cr <sub>2</sub> O <sub>3</sub>	0.52	0.32	0.20	0.51	0.19	0.51	0.20	0.51	0.20					
F	0.67	0.14#	NA	0.30 <sup>#</sup>	NA	0.33 <sup>\$</sup>	NA	0.37#	NA					
Fe <sub>2</sub> O <sub>3</sub>	5.90	8.94	8.97	6.87	6.21	6.49	5.96	6.00	5.6					
K <sub>2</sub> O	0.14	0.24	0.24	0.20	0.24	0.20	0.24	0.18	0.23					
La <sub>2</sub> O <sub>3</sub>	§	0.18	NA	0.05	NA	0.03	NA	< 0.01	NA					
Li <sub>2</sub> O	3.57	3.46**	3.46	3.38**	3.69	3.45**	3.73	3.56**	3.85					
MgO	0.12	0.48	0.43	0.33	0.43	0.36	0.43	0.33	0.43					
MnO	§	0.72	0.21	0.20	0.21	0.14	0.16	0.04	0.05					
Na <sub>2</sub> O	9.58	11.04	9.95	9.84	9.06	9.62	8.96	9.87	8.64					
Nd <sub>2</sub> O <sub>3</sub>	§	0.12	NA	0.03	NA	0.02	NA	< 0.01	NA					
NiO	0.40	0.32	0.26	0.34	0.26	0.31	0.25	0.32	0.26					
P <sub>2</sub> O5	1.05	0.52	0.90	1.00	0.89	1.07	0.98	1.15	1.01					
PbO	0.41	0.16	0.24	0.32	0.40	0.31	0.42	0.33	0.44					
SiO <sub>2</sub>	27.00	39.69	40.59	30.10	31.24	29.35	30.01	27.01	28.77					
SO <sub>3</sub>	0.20	0.08	NA	0.14	NA	0.12	NA	0.13	NA					
SrO	§	0.21	0.27	0.06	0.07	0.04	0.04	0.01	< 0.01					
TiO <sub>2</sub>	0.01	0.15	0.08	0.07	0.08	0.07	0.07	0.05	0.05					
ZnO	0.08	1.10	1.14	0.34	0.28	0.25	0.21	0.10	0.08					
ZrO <sub>2</sub>	0.39	1.74	2.06	0.83	0.93	0.70	0.81	0.50	0.58					
Sum	100.00	100.00	98.11	100.00	95.81	100.00	95.06	100.00	96.67					

§ - Not a target constituent;

NA - Not analyzed

\*\*- for XRF-analyzed compositions, values for B<sub>2</sub>O<sub>3</sub> and Li<sub>2</sub>O were calculated based on simple well-stirred tank model and DCP analyzed values for the first discharge.

# - Fluorine measured by XRF on polished samples

\$ - Fluorine values calculated by interpolation

ORP-56292 Rev. 0

The Catholic University of America Vitreous State Laboratory Melt Rate Enhancement for High Aluminum HLW Glass Formulations Final Report, VSL-08R1360-1, Rev. 0

Test		1	2	3	4	5	6	7	8	
Glass Sample	es	BLX- G-84C	BLX-G- 115A	BLY-G-14B	BLY- G-36B	BLY-G- 80C	BLY-G- 114A	BLZ-G-5B	BLZ-G- 37A	DWPF-EA
	В	10.15	13.18	13.95	12.96	16.92	17.89	29.58	36.86	16695
7-Day PCT Concentration in	Li	6.53	6.79	5.71	5.34	7.32	7.56	9.14	10.45	9565
mg/L	Na	21.89	20.70	17.74	16.56	19.68	20.25	31.04	35.97	13346
ing/L	Si	23.49	21.28	24.33	21.56	23.11	24.31	26.46	27.56	3920
	В	0.22	0.28	0.26	0.24	0.31	0.33	0.51	0.62	$\land$
7-Day PCT	Li	0.43	0.42	0.35	0.33	0.46	0.47	0.56	0.63	
Normalized Concentrations,	Na	0.29	0.27	0.25	0.26	0.30	0.30	0.43	0.51	
g/L	Si	0.15	0.14	0.17	0.15	0.17	0.18	0.20	0.21	
8,	pН	9.92	9.68	9.41	9.33	9.45	9.42	9.40	9.37	
	В	0.11	0.14	0.13	0.12	0.15	0.16	0.25	0.31	
7-Day PCT Normalized Mass	Li	0.22	0.21	0.18	0.17	0.23	0.24	0.28	0.32	
Loss $(g/m^2)$	Na	0.14	0.13	0.13	0.13	0.15	0.15	0.21	0.25	
1000 (g/m )	Si	0.08	0.07	0.08	0.08	0.08	0.09	0.10	0.11	
	В	0.02	0.02	0.02	0.02	0.02	0.02	0.04	0.04	
7-Day PCT Normalized Loss	Li	0.03	0.03	0.03	0.02	0.03	0.03	0.04	0.05	
Rate, $g/d/m^2$	Na	0.02	0.02	0.02	0.02	0.02	0.02	0.03	0.04	] \
reace, g/u/m	Si	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.02	

#### Table 6.9. PCT Results for Melter Glasses (ASTM C1285, 7-days at 90°C, Stainless Steel Vessel; S/V=2000m<sup>-1</sup>).

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	Sample I.D.	Ba	Cd	Cr	Ni	Pb	Zn
Test	UTS Limits #	21	0.11	0.60	11.00	0.75	4.3
1050	Delisting Limits	100	0.48	4.95	22.6	5.00	225
1	BLX-G-84C	0.15	0.04	0.04	0.10	0.11	0.17
2	BLX-G-115A	0.15	0.04	0.05	0.11	0.21	0.14
3	BLY-G-14B	0.18	0.05	0.05	0.08	0.21	0.12
4	BLY-G-36B	0.18	0.05	0.04	0.09	0.27	0.11
5	BLY-G-80C	0.18	0.04	0.10	0.05	0.27	0.12
6	BLY-G-114A	0.17	0.04	0.05	0.12	0.11	0.10
7	BLZ-G-5B	0.26	< 0.03	0.07	0.10	0.55	0.11
8	BLZ-G-37A	0.20	0.03	0.08	0.13	0.39	0.08

 Table 6.10. TCLP Results for Melter Glasses (mg/L).

# For comparison only; does not apply to WTP glasses

Melt Rate Enhancement for High Aluminum HLW Glass Formulations Final Report, VSL-08R1360-1, Rev. 0

 Table 6.11. Glass Pool Samples and Secondary Phase Observations.

	Test	T (°C)	Sampling Date	Sample I.D.	Visual Observations Upon Sampling
1	Before Test	1200	2/4/08	BLX-D-50B	No Secondary Phase
1	End of Test	1200	2/6/08	BLX-D-88A	No Secondary Phase
2	End of Test	1150	2/9/08	BLX-D-119A	No Secondary Phase
3	Before Test	1200	5/3/08	BLX-D-125A	No Secondary Phase
3	End of Test	1200	5/7/08	BLY-D-14A	No Secondary Phase
4	End of Test	1150	5/9/08	BLY-D-36A	No Secondary Phase
5	Before Test	1200	6/2/08	BLY-D-46A	No Secondary Phase
5	End of Test	1200	6/4/08	BLY-D-84A	No Secondary Phase
6	End of Test	1150	6/7/08	BLY-D-114A	No Secondary Phase
7	End of Test	1200	6/25/08	BLZ-D-5A	No Secondary Phase
8	End of Test	1150	6/28/08	BLZ-D-40A	No Secondary Phase

						1			
$\backslash$			Test				-	st 2	
	$\backslash$		02/06/08 09:2	02/08/08 12:22 - 13:22					
	$\mathbf{i}$		5.0% Isokinetic,	13.1% Moist	ure		.2% Isokinet	ic, 8.5% Moi	sture
	$\backslash$	Feed	Emissions			Feed	Emissions		
	$\backslash$	Rate <sup>#</sup>	Rate	% Feed	DF	Rate <sup>#</sup>	Rate	% Feed	DF
		(mg/min)	(mg/min)			(mg/min)	(mg/min)		
	Total <sup>\$</sup>	110500	318	0.29	348	64440	64.7	0.10	995
	Al	11413	16.5	0.14	692	6657	2.06	0.03	3235
	В	4243	17.6	0.41	242	2475	2.60	0.11	951
	Ba	40	0.25	0.63	158	24	0.27	1.15	86.8
	Bi	460	3.51	0.76	131	268	0.58	0.21	467
	Ca	3911	8.50	0.22	460	2282	1.35	0.06	1691
	Cd	16	0.39	2.39	41.9	9	< 0.10	< 1.06	> 94
	Cl*	0	0.46	NC	NC	0	< 0.10	NC	NC
	Cr	320	3.25	1.01	98.6	187	0.98	0.52	191
tte	F*	603	29.1	4.82	20.8	352	6.87	1.95	51.2
sul	Fe	3713	14.5	0.39	256	2166	2.23	0.10	972
Particulate	Κ	105	2.30	2.20	45.4	61	0.52	0.86	117
Pa	Li	1493	6.91	0.46	216	871	1.44	0.17	603
	Mg	65	0.64	0.99	101	38	0.11	0.30	336
	Na	6397	38.5	0.60	166	3732	9.08	0.24	411
	Ni	283	0.86	0.30	330	165	< 0.10	< 0.06	> 1650
	Р	413	0.85	0.21	487	241	0.12	0.05	2008
	Pb	343	2.38	0.69	144	200	0.55	0.28	361
	S*	72	4.74	6.58	15.2	42	2.36	5.62	17.8
	Si	12832	19.3	0.15	664	7485	3.89	0.05	1926
	Zn	58	0.48	0.83	121	34	0.22	0.66	152
	Zr	260	0.52	0.20	497	152	< 0.10	< 0.07	> 1516
	В	4243	53.9	1.27	78.8	2475	26.8	1.08	92.2
Gas	Cl	0	8.83	NC	NC	0	3.76	NC	NC
Ű	F	603	151	25.0	4.0	352	117	33.4	3.0
	S	72	25.4	35.3	2.8	42	9.80	23.3	4.3

#### Table 7.1. Results from DM100 Off-Gas Emission Samples.

<sup>§</sup> - From gravimetric analysis of filters and particulate nitric acid rinses
 <sup>#</sup> - Feed rate calculated from target composition and steady state production rate

\* - Calculated from direct analysis of nitric acid rinses

NA – Not Available

			Test	3			Те	st 4	
			05/07/08 09:4	05/09/08 10:43 - 11:43					
	$\backslash$	105.8% Isokinetic, 13.6% Moisture				100		c, 10.3% Mo	isture
	$\backslash$	Feed	Emissions			Feed	Emissions	, 1010 / 0 1010	
		Rate <sup>#</sup>	Rate	% Feed	DF	Rate <sup>#</sup>	Rate	% Feed	DF
	$\backslash$	(mg/min)	(mg/min)			(mg/min)	(mg/min)		
	Total <sup>\$</sup>	122600	172	0.14	712	83200	32.6	0.04	2550
	Al	12915	7.88	0.06	1638	8764	0.73	0.01	12037
	В	5778	8.40	0.15	688	3920	0.54	0.01	7197
	Ba	47	0.16	0.34	296	32	0.11	0.36	279
	Bi	523	3.35	0.64	156	355	0.35	0.10	1005
	Ca	4421	2.09	0.05	2116	3000	0.31	0.01	9773
	Cd	19	0.36	1.90	52.7	13	< 0.10	< 0.78	> 128
	Cl*	0	19.0	NC	NC	0	12.5	NC	NC
	Cr	366	3.11	0.85	118	249	0.54	0.22	462
ate	F*	683	18.6	2.72	36.8	463	7.33	1.58	63.2
Particulate	Fe	4200	4.03	0.10	1042	2850	0.67	0.02	4243
rtic	K	122	1.23	1.00	99.6	83	0.25	0.31	328
Pa	Li	1688	3.74	0.22	451	1145	0.69	0.06	1651
	Mg	70	0.24	0.34	293	47	< 0.10	< 0.21	> 473
	Na	7237	21.7	0.30	334	4911	5.02	0.10	978
	Ni	322	0.12	0.04	2791	218	< 0.10	< 0.05	> 2184
	Р	468	0.15	0.03	3202	317	< 0.10	< 0.03	> 3173
	Pb	390	2.15	0.55	181	265	0.37	0.14	712
	S*	80	6.23	7.79	12.8	54	2.00	3.70	27
	Si	14519	6.27	0.04	2317	9852	1.10	0.01	8970
	Zn	67	0.31	0.46	218	46	0.27	0.58	171
	Zr	295	0.24	0.08	1221	200	< 0.10	< 0.05	> 2004
	В	5778	82.3	1.42	70.2	3920	19.1	0.49	205
Gas	Cl	0	8.37	NC	NC	0	6.26	NC	NC
9	F	683	206	30.2	3.3	463	88.4	19.1	5.2
\$	S	80	38.6	48.3	2.1	54	16.6	30.6	3.3

#### Table 7.1. Results from DM100 Off-Gas Emission Samples (continued).

<sup>s</sup> - From gravimetric analysis of filters and particulate nitric acid rinses
 <sup>#</sup> - Feed rate calculated from target composition and steady state production rate

\* - Calculated from direct analysis of nitric acid rinses

			Test	5			Та	ort 6	
			06/04/08 11:2	Test 6 06/06/08 12:50 - 13:50					
	$\backslash$	102				06			tura
	$\backslash$		2.8% Isokinetic,	14.5% MOIS	ure		9% Isokineti	c, 8.0% MOIS	sture
		Feed Rate <sup>#</sup>	Emissions	0/ E 1	DE	Feed Rate <sup>#</sup>	Emissions	0/1	DE
	$\backslash$		Rate	% Feed	DF		Rate	% Feed	DF
	Total <sup>\$</sup>	(mg/min) 108900	(mg/min) 144	0.13	756	(mg/min) 58700	(mg/min) 40.3	0.07	1457
-	Al	11992	3.11	0.13	3853		0.74		8715
-	B					6457		0.01	
-		5365	7.37	0.14	728	2889	1.65	0.06	1751
-	Ba	44	< 0.10	< 0.23	> 437	24	< 0.10	< 0.43	> 235
-	Bi	485	2.37	0.49	204	261	0.35	0.13	742
	Ca	4105	2.02	0.05	2037	2210	0.49	0.02	4481
	Cd	17	0.20	1.12	89.1	9	< 0.10	< 1.06	> 94
	Cl*	0	13.8	NC	NC	0	9.37	NC	NC
	Cr	340	2.47	0.73	138	183	0.68	0.37	269
ate	F*	634	19.9	3.14	31.9	341	5.53	1.62	61.7
Sul	Fe	3900	4.47	0.11	873	2100	1.16	0.06	1818
Particulate	K	113	0.95	0.84	119	61	0.24	0.39	258
Pa	Li	1567	3.57	0.23	439	844	0.94	0.11	899
	Mg	65	0.16	0.25	398	35	< 0.10	< 0.29	> 348
	Na	6720	18.8	0.28	357	3619	5.45	0.15	664
	Ni	299	< 0.10	< 0.03	> 2988	161	< 0.10	< 0.06	> 1609
	Р	434	0.10	0.02	4190	234	< 0.10	< 0.04	> 2338
	Pb	362	1.48	0.41	245	195	0.33	0.17	591
	S*	74	6.08	8.22	12.2	40	1.84	4.60	21.7
	Si	13482	4.07	0.03	3316	7260	1.01	0.01	7173
	Zn	63	0.20	0.31	320	34	0.12	0.36	280
	Zr	274	0.24	0.09	1136	148	< 0.10	< 0.07	> 1477
	В	5365	65.3	1.22	82.1	2889	19.7	0.68	147
Gas	Cl	0	7.94	NC	NC	0	5.95	NC	NC
Ö	F	634	206	32.6	3.1	341	81.0	23.7	4.2
	S	74	26.1	35.2	2.8	40	14.0	35.1	2.8

#### Table 7.1. Results from DM100 Off-Gas Emission Samples (continued).

<sup>s</sup> - From gravimetric analysis of filters and particulate nitric acid rinses
 <sup>#</sup> - Feed rate calculated from target composition and steady state production rate

\* - Calculated from direct analysis of nitric acid rinses

$\square$			Test	7			Те	st 8	
	\		6/24/08 19:4		06/27/08/08 17:26 - 18:26				
	$\backslash$	98.5% Isokinetic, 14.6% Moisture					5% Isokineti		
	$\backslash$	Feed	Emissions			Feed	Emissions	,	
		Rate <sup>#</sup>	Rate	% Feed	DF	Rate <sup>#</sup>	Rate	% Feed	DF
	$\backslash$	(mg/min)	(mg/min)			(mg/min)	(mg/min)		
	Total <sup>\$</sup>	138200	171	0.12	806	87500	45.2	0.05	1938
	Al	14266	13.8	0.10	1038	9035	2.77	0.03	3261
	В	6700	8.58	0.13	781	4243	1.99	0.05	2127
	Ba	50	< 0.10	< 0.20	> 504	32	< 0.10	< 0.31	> 319
	Bi	575	2.56	0.44	225	364	0.50	0.14	726
	Ca	4487	2.42	0.05	1852	2842	0.47	0.02	6086
	Cd	20	0.14	0.71	141	13	< 0.10	< 0.78	> 128
	Cl*	0	0.21	NC	NC	0	8.79	NC	NC
ļſ	Cr	400	1.94	0.48	207	254	0.77	0.30	329
lte	F*	754	19.3	2.56	39.1	477	5.53	1.16	86.3
sul	Fe	4641	4.09	0.09	1136	2939	0.91	0.03	3235
Particulate	K	131	1.19	0.91	110	83	0.40	0.48	208
Pa	Li	1866	3.48	0.19	537	1182	1.11	0.09	1065
ļſ	Mg	81	0.22	0.27	369	52	< 0.10	< 0.19	> 516
ļſ	Na	7996	20.1	0.25	399	5064	7.38	0.15	686
	Ni	354	0.29	0.08	1219	224	< 0.10	< 0.04	> 2240
	Р	516	0.41	0.08	1258	327	0.38	0.12	857
	Pb	428	1.08	0.25	395	271	0.39	0.14	701
	S*	90	4.41	4.90	20.4	57	1.60	2.81	35.6
	Si	14199	6.42	0.05	2211	8993	1.56	0.02	5776
	Zn	72	0.10	0.14	700	46	< 0.10	< 0.22	> 458
	Zr	325	0.15	0.05	2209	206	< 0.10	< 0.05	> 2057
	В	6700	55.2	0.82	121	4243	24.1	0.57	176
Gas	Cl	0	14.8	NC	NC	0	8.76	NC	NC
IJ	F	754	175	23.2	4.3	477	99.9	20.9	4.8
	S § Errorre o	90	40.9	45.4	2.2	57	15.4	27.0	3.7

#### Table 7.1. Results from DM100 Off-Gas Emission Samples (continued).

<sup>\$</sup> - From gravimetric analysis of filters and particulate nitric acid rinses
 <sup>#</sup> - Feed rate calculated from target composition and steady state production rate

\* - Calculated from direct analysis of nitric acid rinses

$\begin{tabular}{ c c c c c c c c c c c c c c c c c c c$				Test	1			Та	~ <b>t</b> )		
$\begin{array}{ c c c c c c c c c c c c c c c c c c c$	$\backslash$						Test 2				
$\begin{array}{ c c c c c c c c c c c c c c c c c c c$		$\backslash$	1.00				06			,	
$\begin{tabular}{ c c c c c c c c c c c c c c c c c c c$				/	39.4% Moist	ure			c, 29.2% MOI	sture	
Image: mark text         (mg/min)         (mg/min)         (mg/min)         (mg/min)         (mg/min)           Image: mark text         Total <sup>8</sup> 1569000         2453         0.16         640         1109000         1114         0.10         995           Al         158508         116         0.07         1367         110955         50.5         0.05         2194           B         58927         137         0.23         430         41249         59.9         0.15         689           Ba         560         2.65         0.47         211         392         0.69         0.18         564           Bi         6391         26.6         0.42         240         4474         13.1         0.29         341           Ca         54324         53.5         0.10         1016         38027         22.7         0.06         1675           Cd         224         2.19         0.98         102         157         0.70         0.44         225           Cl*         0         14.1         NC         NC         0         9.50         NC         NC           F*         8375         332         3.97         25.2					0/ 5 1	DE			0/ E 1	DE	
$ \begin{array}{ c c c c c c c c c c c c c c c c c c c$					% Feed	DF			% Feed	DF	
Al         158508         116         0.07         1367         110955         50.5         0.05         2194           B         58927         137         0.23         430         41249         59.9         0.15         689           Ba         560         2.65         0.47         211         392         0.69         0.18         564           Bi         6391         26.6         0.42         240         4474         13.1         0.29         341           Ca         54324         53.5         0.10         1016         38027         22.7         0.06         1675           Cd         224         2.19         0.98         102         157         0.70         0.444         225           Cl*         0         14.1         NC         NC         0         9.50         NC         NC           Cr*         4447         24.7         0.55         180         3113         15.1         0.48         207           F*         8375         332         3.97         25.2         5863         211         3.60         27.8           Fe         51570         111         0.21         466         360		T ( 18			0.16	(10			0.10	005	
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$\begin{array}{ c c c c c c c c c c c c c c c c c c c$											
Ca         54324         53.5         0.10         1016         38027         22.7         0.06         1675           Cd         224         2.19         0.98         102         157         0.70         0.44         225           Cl*         0         14.1         NC         NC         0         9.50         NC         NC           Cr         4447         24.7         0.55         180         3113         15.1         0.48         207           F*         8375         332         3.97         25.2         5863         211         3.60         27.8           Fe         51570         111         0.21         466         36099         49.0         0.14         737           K         1453         16.2         1.11         89.7         1017         8.47         0.83         120           Li         20729         46.3         0.22         448         14511         21.8         0.15         667           Mg         904         5.61         0.62         161         633         2.27         0.36         278.4           Na         88847         310         0.35         287         62193											
Cd         224         2.19         0.98         102         157         0.70         0.44         225           Cl*         0         14.1         NC         NC         0         9.50         NC         NC           Cr         4447         24.7         0.55         180         3113         15.1         0.48         207           F*         8375         332         3.97         25.2         5863         211         3.60         27.8           Fe         51570         111         0.21         466         36099         49.0         0.14         737           K         1453         16.2         1.11         89.7         1017         8.47         0.83         120           Li         20729         46.3         0.22         448         14511         21.8         0.15         667           Mg         904         5.61         0.62         161         633         2.27         0.36         278.4           Na         88847         310         0.35         287         62193         141         0.23         441           Ni         3929         4.45         0.11         883         2750											
Cl*         0         14.1         NC         NC         0         9.50         NC         NC           Cr         4447         24.7         0.55         180         3113         15.1         0.48         207           F*         8375         332         3.97         25.2         5863         211         3.60         27.8           Fe         51570         111         0.21         466         36099         49.0         0.14         737           K         1453         16.2         1.11         89.7         1017         8.47         0.83         120           Li         20729         46.3         0.22         448         14511         21.8         0.15         667           Mg         904         5.61         0.62         161         633         2.27         0.36         278.4           Na         88847         310         0.35         287         62193         141         0.23         441           Ni         3929         4.45         0.11         883         2750         2.14         0.08         1285           P         5731         7.41         0.13         773         4011 <td></td> <td></td> <td>54324</td> <td></td> <td>0.10</td> <td>1016</td> <td>38027</td> <td>22.7</td> <td>0.06</td> <td></td>			54324		0.10	1016	38027	22.7	0.06		
$\begin{array}{ c c c c c c c c c c c c c c c c c c c$		Cd	224	2.19	0.98		157	0.70	0.44	225	
$\begin{array}{c c c c c c c c c c c c c c c c c c c $		Cl*	0	14.1	NC	NC	0	9.50	NC	NC	
$ \begin{array}{c c c c c c c c c c c c c c c c c c c $		Cr	4447	24.7	0.55	180	3113	15.1	0.48	207	
$ \begin{array}{ c c c c c c c c c c c c c c c c c c c$	tte	F*	8375	332	3.97	25.2	5863	211	3.60	27.8	
$ \begin{array}{ c c c c c c c c c c c c c c c c c c c$	ula	Fe	51570	111	0.21	466	36099	49.0	0.14	737	
$ \begin{array}{ c c c c c c c c c c c c c c c c c c c$	rtic	K	1453	16.2	1.11	89.7	1017	8.47	0.83	120	
$ \begin{array}{ c c c c c c c c c c c c c c c c c c c$	Pa	Li	20729	46.3	0.22	448	14511	21.8	0.15	667	
Ni         3929         4.45         0.11         883         2750         2.14         0.08         1285           P         5731         7.41         0.13         773         4011         3.27         0.08         1226           Pb         4758         15.4         0.32         308         3330         8.91         0.27         374           S*         1002         79.5         7.93         12.6         701         27.3         3.89         25.7           Si         178221         130         0.07         1368         124755         60.1         0.05         2074           Zn         803         3.01         0.37         267         562         1.15         0.20         489           Zr         3609         2.79         0.08         1294         2526         1.12         0.04         2263           B         58927         846         1.44         69.7         41249         441.8         1.07         93.4           S         Cl         0         168         NC         NC         0         83.6         NC         NC		Mg	904	5.61	0.62	161	633	2.27	0.36	278.4	
P         5731         7.41         0.13         773         4011         3.27         0.08         1226           Pb         4758         15.4         0.32         308         3330         8.91         0.27         374           S*         1002         79.5         7.93         12.6         701         27.3         3.89         25.7           Si         178221         130         0.07         1368         124755         60.1         0.05         2074           Zn         803         3.01         0.37         267         562         1.15         0.20         489           Zr         3609         2.79         0.08         1294         2526         1.12         0.04         2263           B         58927         846         1.44         69.7         41249         441.8         1.07         93.4           G         Cl         0         168         NC         NC         0         83.6         NC         NC		Na	88847	310	0.35	287	62193	141	0.23	441	
Pb         4758         15.4         0.32         308         3330         8.91         0.27         374           S*         1002         79.5         7.93         12.6         701         27.3         3.89         25.7           Si         178221         130         0.07         1368         124755         60.1         0.05         2074           Zn         803         3.01         0.37         267         562         1.15         0.20         489           Zr         3609         2.79         0.08         1294         2526         1.12         0.04         2263           B         58927         846         1.44         69.7         41249         441.8         1.07         93.4           Cl         0         168         NC         NC         0         83.6         NC         NC		Ni	3929	4.45	0.11	883	2750	2.14	0.08	1285	
S*         1002         79.5         7.93         12.6         701         27.3         3.89         25.7           Si         178221         130         0.07         1368         124755         60.1         0.05         2074           Zn         803         3.01         0.37         267         562         1.15         0.20         489           Zr         3609         2.79         0.08         1294         2526         1.12         0.04         2263           B         58927         846         1.44         69.7         41249         441.8         1.07         93.4           Cl         0         168         NC         NC         0         83.6         NC         NC		Р	5731	7.41	0.13	773	4011	3.27	0.08	1226	
Si         178221         130         0.07         1368         124755         60.1         0.05         2074           Zn         803         3.01         0.37         267         562         1.15         0.20         489           Zr         3609         2.79         0.08         1294         2526         1.12         0.04         2263           B         58927         846         1.44         69.7         41249         441.8         1.07         93.4           S         Cl         0         168         NC         NC         0         83.6         NC         NC		Pb	4758	15.4	0.32	308	3330	8.91	0.27	374	
Zn         803         3.01         0.37         267         562         1.15         0.20         489           Zr         3609         2.79         0.08         1294         2526         1.12         0.04         2263           B         58927         846         1.44         69.7         41249         441.8         1.07         93.4           G         Cl         0         168         NC         NC         0         83.6         NC         NC		S*	1002	79.5	7.93	12.6	701	27.3	3.89	25.7	
Zr         3609         2.79         0.08         1294         2526         1.12         0.04         2263           B         58927         846         1.44         69.7         41249         441.8         1.07         93.4           G         Cl         0         168         NC         NC         0         83.6         NC         NC		Si	178221	130	0.07	1368	124755	60.1	0.05	2074	
B         58927         846         1.44         69.7         41249         441.8         1.07         93.4           g         Cl         0         168         NC         NC         0         83.6         NC         NC		Zn	803	3.01	0.37	267	562	1.15	0.20	489	
B         58927         846         1.44         69.7         41249         441.8         1.07         93.4           g         Cl         0         168         NC         NC         0         83.6         NC         NC	[	Zr	3609	2.79	0.08	1294	2526	1.12	0.04	2263	
		В	58927	846	1.44	69.7	41249	441.8	1.07	93.4	
	as	Cl	0	168	NC	NC	0	83.6	NC	NC	
\[\  F   8375   1914   22.9   4.4   5863   1091   18.6   5.4	Ű	F	8375	1914	22.9	4.4	5863	1091	18.6	5.4	
S 1002 241 24.1 4.2 701 181 25.8 3.9		S	1002	241	24.1		701	181			

#### Table 7.2. Results from DM1200 Off-Gas Emission Samples.

<sup>s</sup> - From gravimetric analysis of filters and particulate nitric acid rinses
 <sup>#</sup> - Feed rate calculated from target composition and steady state production rate

\* - Calculated from direct analysis of nitric acid rinses

<u> </u>					
$\backslash$	、 、		Test		
	$\backslash$	(0)	08/14/08 13:		
			.6% Isokinetic, 2	26.9% Moisti	ire
		Feed	Emissions	0/1	DE
	$\backslash$	Rate <sup>#</sup>	Rate	% Feed	DF
	Total <sup>\$</sup>	(mg/min)	(mg/min)	0.21	477
		1120500	2351	0.21	477
	Al	110955	126		881
	B	41249	152	0.37	271
	Ba	392	1.55	0.40	253
	Bi	4474	27.9	0.62	160
	Ca	38027	59.7	0.16	637
	Cd	157	1.09	0.69	144
	Cl*	0	5.23	NC	NC
	Cr	3113	19.2	0.62	162
Particulate	F*	5863	351	5.99	16.7
cul	Fe	36099	109	0.30	330
urti	K	1017	14.6	1.43	69.9
$\mathbf{P}_{\mathbf{\hat{c}}}$	Li	14511	38.3	0.26	379
	Mg	633	4.82	0.76	131
	Na	62193	302	0.49	206
	Ni	2750	4.92	0.18	559
	Р	4011	7.12	0.18	564
	Pb	3330	12.7	0.38	262
	S*	701	50.7	7.24	13.8
	Si	124755	146	0.12	852
	Zn	562	2.20	0.39	255
	Zr	2526	2.47	0.10	1023
	В	41249	539	1.31	76.5
Gas	Cl	0	107	NC	NC
Ö		5863	1459	24.9	4.0
s	S	701	229	32.6	3.1

#### Table 7.2. Results from DM1200 Off-Gas Emission Samples (continued).

<sup>8</sup> - From gravimetric analysis of filters and particulate nitric acid rinses
 <sup>#</sup> - Feed rate calculated from target composition and steady state production rate
 \* - Calculated from direct analysis of nitric acid rinses

NC – Not Calculated

		1		2		3	4		
	Avg.	Range	Avg.	Range	Avg.	Range	Avg.	Range	
N <sub>2</sub> O	<1.0	<1.0 - 1.5	<1.0	<1.0 - 1.2	<1.0	<1.0 - 1.5	<1.0	<1.0 - 1.9	
NO	88.0	<1.0 - 410	51.3	5.2 - 232	84.0	<1.0 - 248	59.9	1.1 - 250	
NO <sub>2</sub>	5.0	<1.0 - 53.5	4.1	<1.0 - 37.3	6.3	<1.0 - 46.4	8.9	<1.0-71.0	
NH <sub>3</sub>	<1.0	NA	<1.0	NA	<1.0	NA	<1.0	NA	
H <sub>2</sub> O [%]	7.0	0.8 - 22.2	4.3	1.3 - 14.5	7.4	2.3 - 19.1	6.3	0.9 - 17.8	
CO <sub>2</sub>	1337	352 - 4952	1078	382 - 3803	1629	92 - 3690	1340	56 - 4179	
Nitrous Acid	<1.0	NA	<1.0	NA	<1.0	NA	<1.0	NA	
Nitric Acid	<1.0	NA	<1.0	NA	<1.0	NA	<1.0	NA	
HCN	<1.0	NA	<1.0	NA	<1.0	NA	<1.0	NA	
$SO_2$	4.1	<1.0 - 8.8	1.9	<1.0 - 7.9	3.4	<1.0 - 9.7	2.3	<1.0-6.1	
СО	<1.0	<1.0 - 2.9	<1.0	<1.0 - 70.1	<1.0	<1.0 - 3.4	<1.0	<1.0 -2.8	
HCl	<1.0	<1.0 - 2.1	<1.0	<1.0 - 1.2	1.0	<1.0 - 2.5	<1.0	<1.0 - 2.4	
HF	13.5	<1.0 - 41.4	17.2	9.6 - 38.4	15.5	1.3 - 30.6	22.1	7.8 - 37.3	

### Table 7.3. Concentrations (ppmv) of Selected Speciesin DM100 Exhaust Measured by FTIR Spectroscopy.

NA : Not applicable.

		5		6		7	8		
	Avg.	Range	Avg.	Range	Avg.	Range	Avg.	Range	
N <sub>2</sub> O	<1.0	<1.0 - 1.0	<1.0	<1.0 - 1.2	<1.0	<1.0 - 2.8	<1.0	<1.0-2.9	
NO	50.4	<1.0 - 270	31.8	<1.0 - 254	86.8	36.7 - 307	55.4	2.2 - 282	
NO <sub>2</sub>	3.2	<1.0 - 54.9	2.5	<1.0 - 73.0	6.9	<1.0 - 51.7	5.4	<1.0 - 52.5	
NH <sub>3</sub>	<1.0	NA	<1.0	<1.0 - 9.2	<1.0	NA	<1.0	NA	
H <sub>2</sub> O [%]	4.6	2.0 - 10.2	4.0	1.8 - 11.0	7.5	3.6 - 16.9	5.6	1.5 - 14.6	
CO <sub>2</sub>	1073	<1.0 - 3402	790	<1.0 - 3860	1836	922 - 5169	1336	369 - 5208	
Nitrous Acid	<1.0	NA	<1.0	NA	<1.0	NA	<1.0	NA	
Nitric Acid	<1.0	NA	<1.0	<1.0 - 3.9	<1.0	NA	<1.0	NA	
HCN	<1.0	NA	<1.0	NA	<1.0	NA	<1.0	NA	
$SO_2$	2.2	<1.0 - 9.0	1.2	<1.0 - 20.2	6.0	1.1 - 14.2	3.6	<1.0 - 16.5	
CO	<1.0	<1.0 - 1.0	<1.0	<1.0 - 1.6	<1.0	<1.0 - 3.2	<1.0	<1.0 - 3.1	
HCl	<1.0	<1.0 - 1.2	<1.0	<1.0 - 2.0	1.5	1.5 - 3.0	<1.0	<1.0 - 1.4	
HF	8.6	2.2 - 15.7	11.9	8.1 - 25.4	29.3	9.5 - 41.9	23.9	17.9 - 34.0	

### Table 7.3. Concentrations (ppmv) of Selected Speciesin DM100 Exhaust Measured by FTIR Spectroscopy (continued).

NA : Not applicable.

	Me	elter out	let	W	ESP out	let	PBS outlet			
	Aver.	Min.	Max.	Aver.	Min.	Max.	Aver.	Min.	Max.	
N <sub>2</sub> O	1.3	0.7	2.0	1.6	<1.0	10.9	1.4	<1.0	1.7	
NO	594	361	751	611	12.3	1303	520	260	626	
NO <sub>2</sub>	68.6	44.1	78.9	120	<1.0	424	106	54.8	136	
NH <sub>3</sub>	< 1.0	< 1.0	< 1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	
H <sub>2</sub> O %	32.9	27.2	35.2	8.5	1.6	16.1	3.6	3.1	3.8	
CO <sub>2</sub> %	0.73	0.47	0.78	0.75	0.01	1.89	0.79	0.42	0.95	
Nitrous Acid	1.7	1.2	2.0	2.4	<1.0	7.2	<1.0	<1.0	<1.0	
Nitric Acid	1.4	0.2	2.0	1.5	<1.0	5.1	<1.0	<1.0	<1.0	
HCN	<1.0	<1.0	<1.0	<1.0	<1.0	3.2	<1.0	<1.0	<1.0	
$SO_2$	5.1	3.3	7.3	<1.0	<1.0	7.0	<1.0	<1.0	<1.0	
СО	2.5	2.0	3.0	2.9	<1.0	15.4	2.0	1.0	3.0	
HCl	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	
HF	14.6	10.9	16.3	4.5	1.0	7.8	2.5	2.3	2.8	

### Table 7.4 Concentrations [ppmv] of Selected Species in DM1200 Test 1 ExhaustMeasured by FTIR Spectroscopy.

	Melter outlet			S	SBS outlet			WESP outlet			PBS outlet		
	Aver.	Min.	Max.	Aver.	Min.	Max.	Aver.	Min.	Max.	Aver.	Min.	Max.	
N <sub>2</sub> O	1.0	<1.0	1.8	2.3	1.6	5.5	1.5	<1.0	4.9	2.0	<1.0	4.3	
NO	459	75	659	561	364	926	397	85	927	399	243	704	
NO <sub>2</sub>	73.6	28.6	130	107	65.9	264	92.7	22.9	321	90.2	36.3	186	
NH <sub>3</sub>	<1.0	<1.0	1.6	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	
H <sub>2</sub> O %	32.3	26.9	53.5	7.9	7.5	8.3	6.9	5.5	11.0	3.2	3.0	4.6	
CO <sub>2</sub> %	0.83	0.62	1.15	0.77	0.56	1.27	0.59	0.23	1.23	0.60	0.39	1.06	
Nitrous Acid	<1.0	<1.0	2.0	1.3	1.0	2.7	1.4	<1.0	4.4	<1.0	<1.0	<1.0	
Nitric Acid	<1.0	<1.0	1.3	<1.0	<1.0	1.8	<1.0	<1.0	2.4	<1.0	<1.0	9.3	
HCN	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	
$SO_2$	18.3	10.6	23.3	7.8	4.4	11.0	4.5	<1.0	9.5	2.6	<1.0	5.9	
СО	5.9	1.5	98.7	5.8	2.4	83.7	3.3	<1.0	22.3	5.0	1.0	140	
HCl	<1.0	<1.0	5.2	<1.0	<1.0	<1.0	<1.0	<1.0	1.2	<1.0	<1.0	15.4	
HF	20.8	9.7	54.8	9.9	9.3	10.3	5.8	2.8	12.2	6.1	4.5	28.4	

## Table 7.5. Concentrations [ppmv] of Selected Species in DM1200 Test 2 ExhaustMeasured by FTIR Spectroscopy.

	Melter outlet		S	BS outle	et	W	ESP out	let	PBS outlet			
	Aver.	Min.	Max.	Aver.	Min.	Max.	Aver.	Min.	Max.	Aver.	Min.	Max.
N <sub>2</sub> O	1.5	1.2	1.7	2.6	1.3	4.2	2.1	<1.0	11.9	2.3	<1.0	6.2
NO	423	276	673	550	235	778	415	118	631	375	<1.0	725
NO <sub>2</sub>	69.0	54.5	77.6	105	41.9	206	94.8	22.8	178	74.8	<1.0	220
NH <sub>3</sub>	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	1.4
H <sub>2</sub> O %	29.9	26.1	35.6	8.1	7.0	8.9	7.2	6.5	18.4	3.1	2.4	4.4
CO <sub>2</sub> %	0.67	0.46	0.86	0.75	0.38	1.16	0.58	<1.0	0.94	0.50	<1.0	1.20
Nitrous Acid	<1.0	<1.0	1.7	1.2	<1.0	2.3	1.4	<1.0	2.6	0.4	<1.0	1.2
Nitric Acid	<1.0	<1.0	1.2	0.8	<1.0	1.6	<1.0	<1.0	1.8	<1.0	<1.0	<1.0
HCN	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0	<1.0
SO <sub>2</sub>	17.1	11.7	22.8	7.5	3.3	11.8	5.9	1.3	11.6	2.1	<1.0	7.0
СО	3.6	2.4	4.6	3.8	1.1	10.3	4.0	<1.0	96.8	3.1	<1.0	32.7
HCl	<1.0	<1.0	1.4	<1.0	<1.0	1.6	<1.0	<1.0	4.9	<1.0	<1.0	4.1
HF	19.9	11.6	33.9	10.7	9.3	13.8	9.1	1.1	17.7	3.9	2.1	6.7

# Table 7.6. Concentrations [ppmv] of Selected Species in DM1200 Test 3 Exhaust Measured by FTIR Spectroscopy.

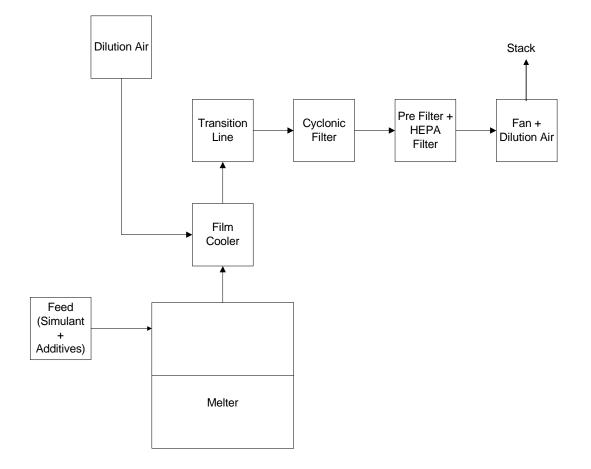


Figure 1.1. Schematic diagram of DuraMelter 100 vitrification system.

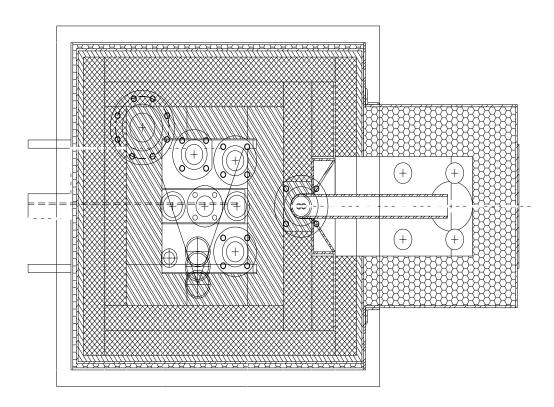


Figure 1.2.a. Schematic diagram showing cross-section through the DM100-BL-melter. Plan view showing locations of lid ports.

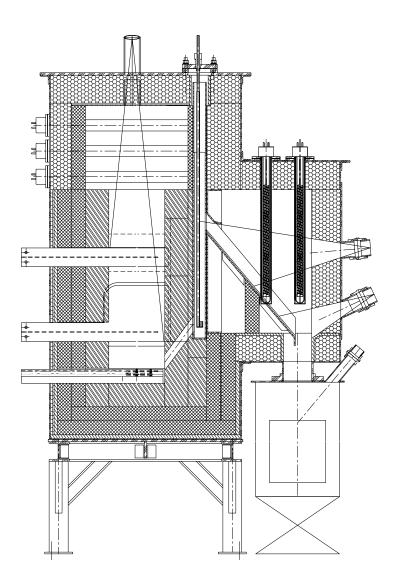
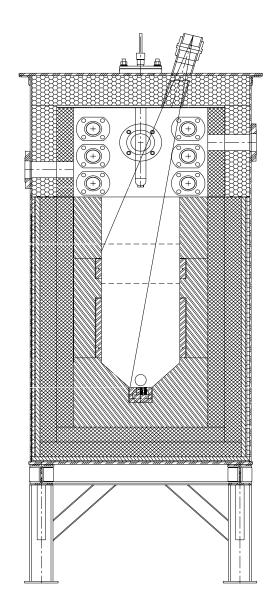
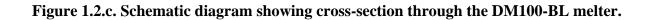


Figure 1.2.b. Schematic diagram showing cross-section through the DM100-BL melter.





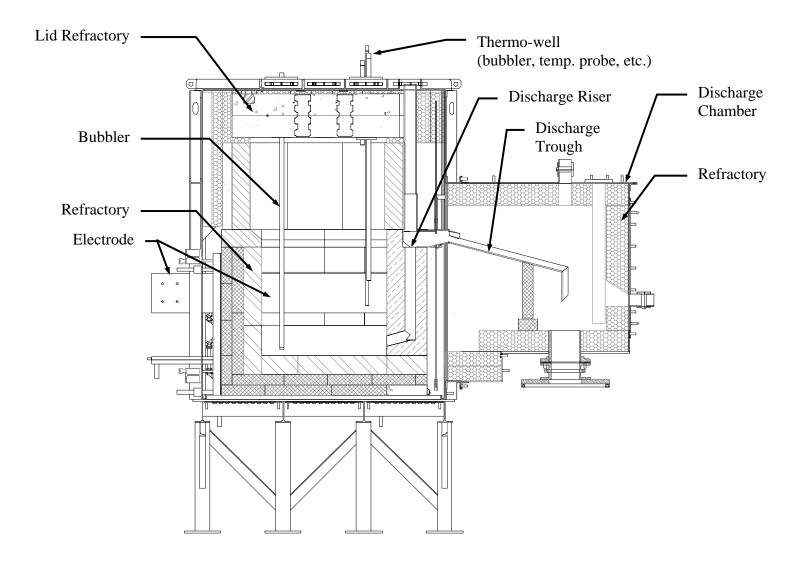


Figure 1.3. Cross-section of the DM1200 melter through the discharge chamber.

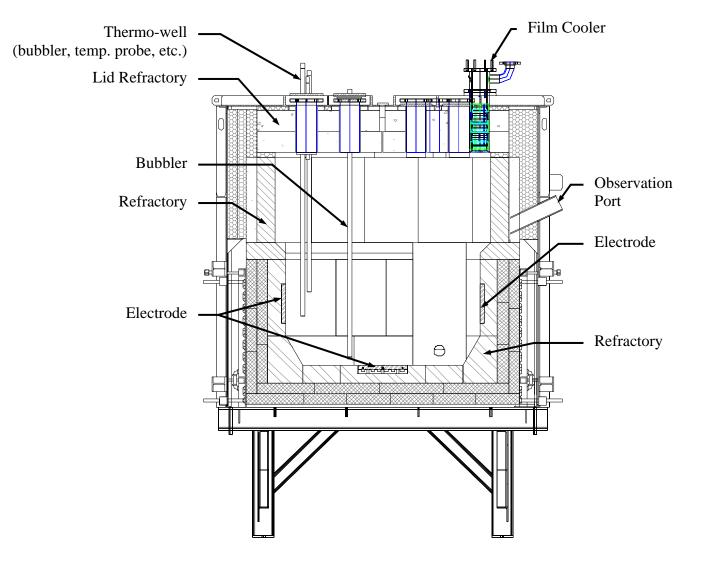


Figure 1.4. Cross-section through the DM1200 melter showing electrodes.

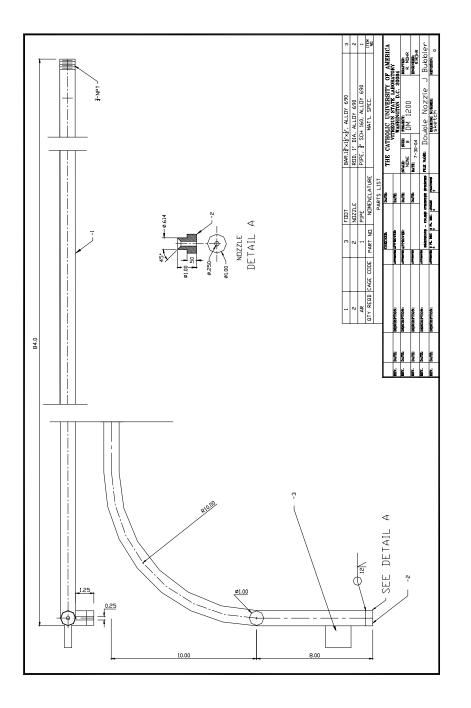
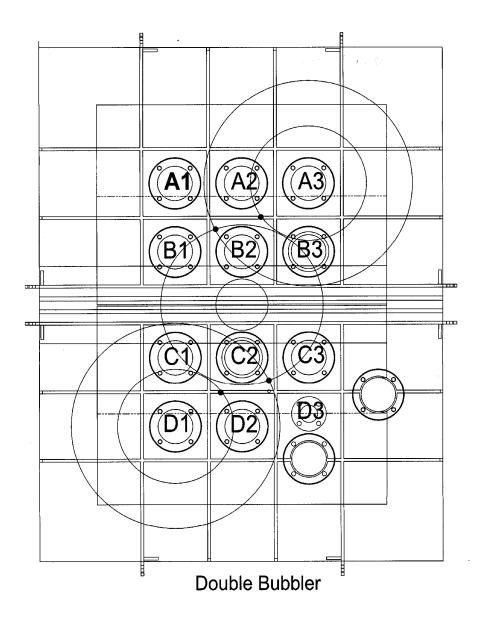
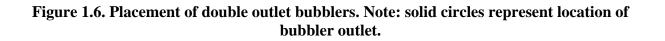


Figure 1.5. Specifications of Double Outlet "J" Bubbler.





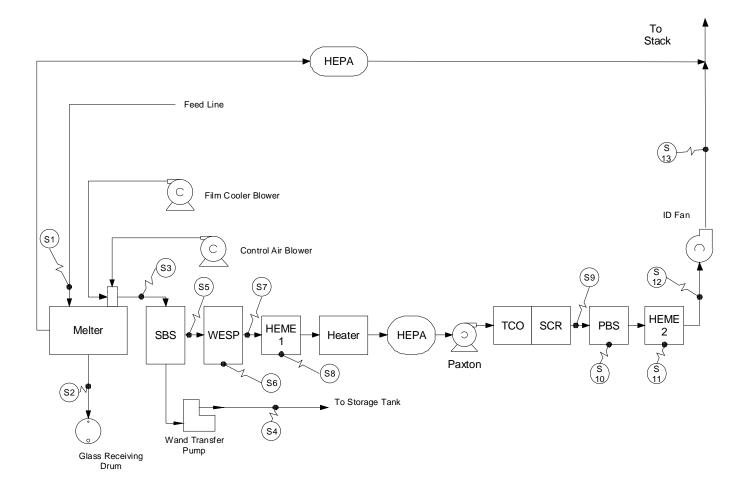


Figure 1.7. Schematic diagram of DM1200 off-gas system. "Sx" indicates sampling point.

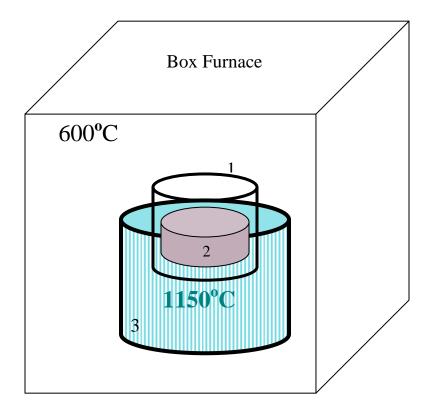


Figure 3.1. Schematic drawing of vertical gradient furnace (VGF) for feed conversion test (1=ceramic crucible half inside the local heater; 2=feed for 20 gram glass; 3=local heater at 1150°C).

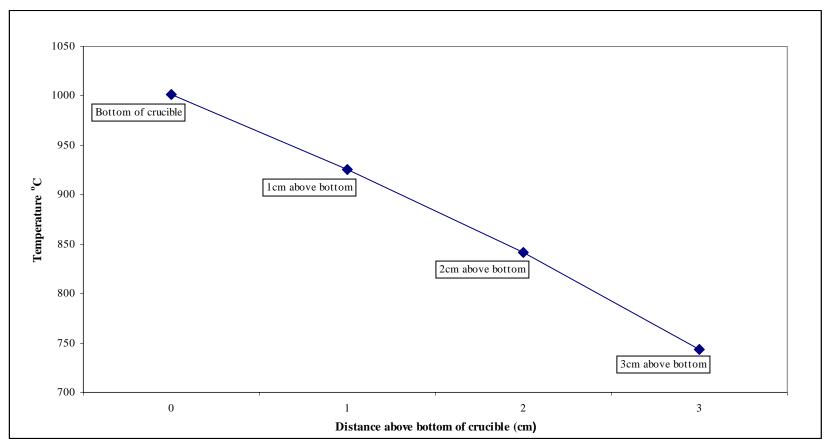


Figure 3.2. Temperature gradient (inside the loaded ceramic crucible) of the Vertical Gradient Furnace (VGF).

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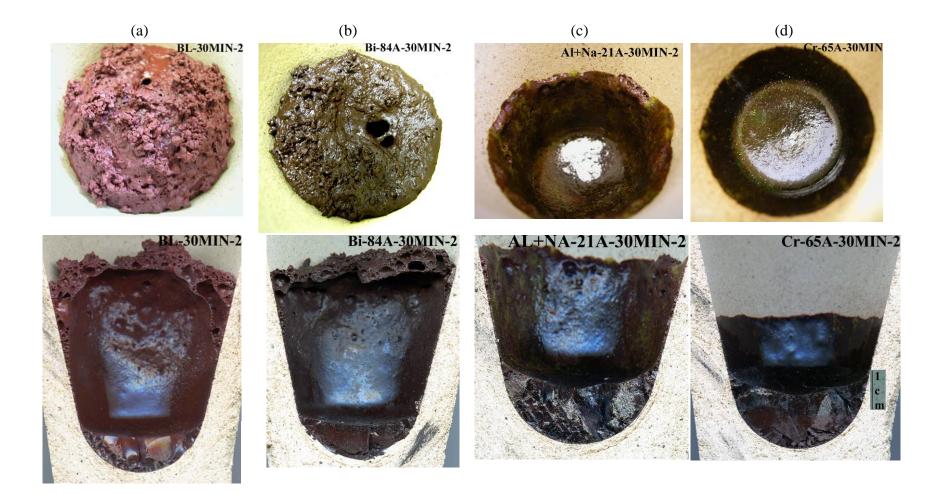


Figure 3.3. Top views and cross sections of six existing melter feeds after VGF tests (30 minutes) (a) HLW-E-Al-27, (b) HLW-E-Bi-6; (c) HLW-E-ANa-22, (d) HLW-E-CrM, (e) Matrix 1-B1, and (f) Matrix 2-9.

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(f)



Figure 3.3. Top views and cross sections of six existing melter feeds after VGF tests (30 minutes) (a) HLW-E-Al-27, (b) HLW-E-Bi-6; (c) HLW-E-ANa-22, (d) HLW-E-CrM, (e) Matrix 1-B1, and (f) Matrix 2-9 (continued).

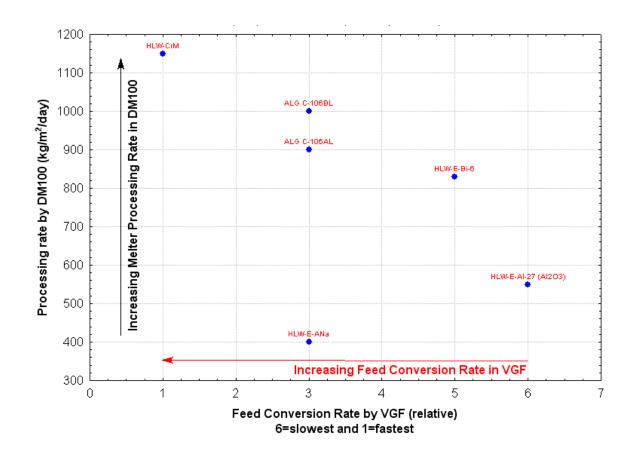


Figure 3.4. Preliminary evaluation of melt rate ranking by VGF as compared to the glass production rate determined by DM100 melter test using the same melter feed. The definition of the VGF melt rate estimation is given in Table 3.5.



(b)

(c)



Figure 3.5. Time evolution of the baseline Al-limited melter feed (HLW-E-Al-27) after VGF tests: (a) 30 minutes, (b) 45 minutes and (c) 60 minutes.

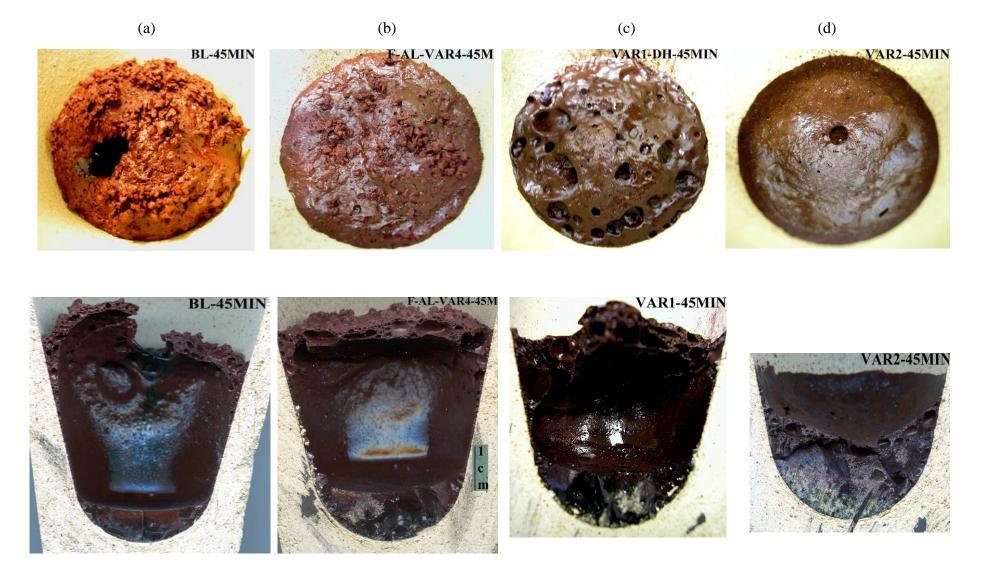


Figure 3.6. Impacts of boron and aluminum sources on melt rate in VGF experiments (45 minutes). (a) Al<sub>2</sub>O<sub>3</sub> and borax, (b) Al(OH)<sub>3</sub> and borax, (c) Al<sub>2</sub>O<sub>3</sub> and boric acid, and (d) Al(OH)<sub>3</sub> and boric acid

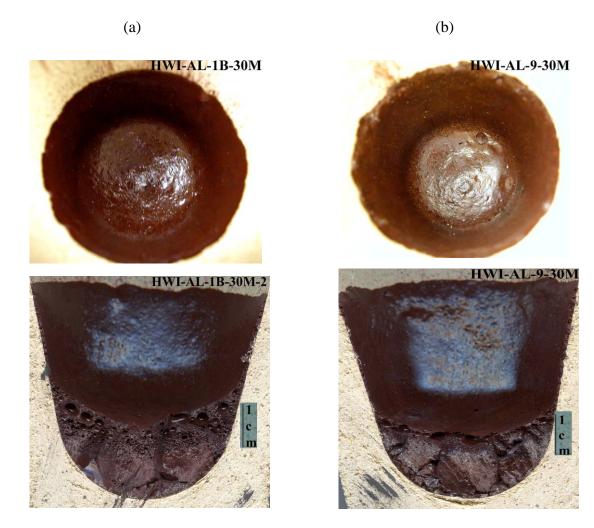


Figure 3.7. Top views and cross sections of Group 1 formulations after 30 minute test in VGF. (a) HWI-Al-1, and (b) HWI-Al-9

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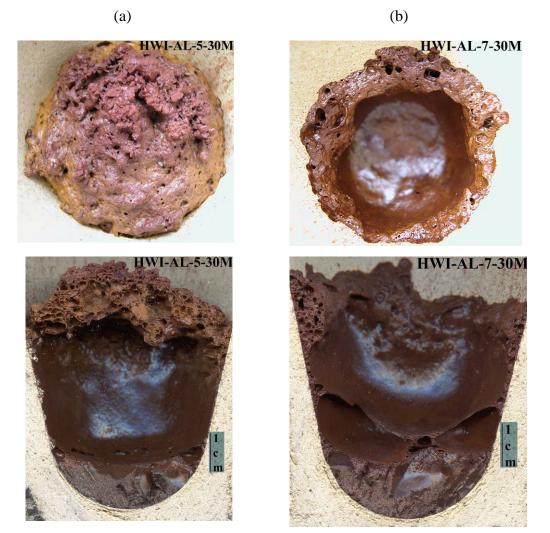


Figure 3.8. Top views and cross sections of Group 3 formulations after 30 minute test in VGF. (a) HWI-Al-5, and (b) HWI-Al-7.

F-18

(a)

(b)

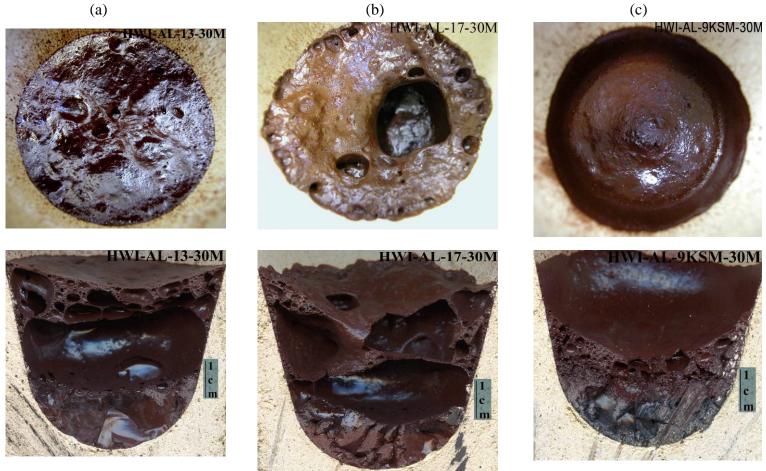


Figure 3.9. Top views and cross sections of Group 4 formulations after 30 minute test in VGF. (a) HWI-Al-13, (b) HWI-Al-17 and (b) HWI-Al-20.

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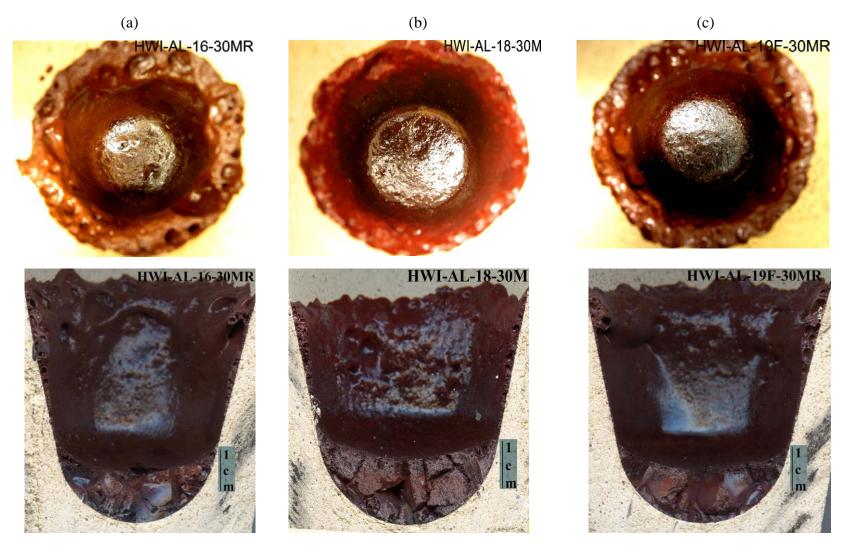


Figure 3.10. Top views and cross sections of Group 5 formulations after 30 minute test in VGF. (a) HWI-Al-16, (b) HWI-Al-18 and (c) HWI-Al-19.

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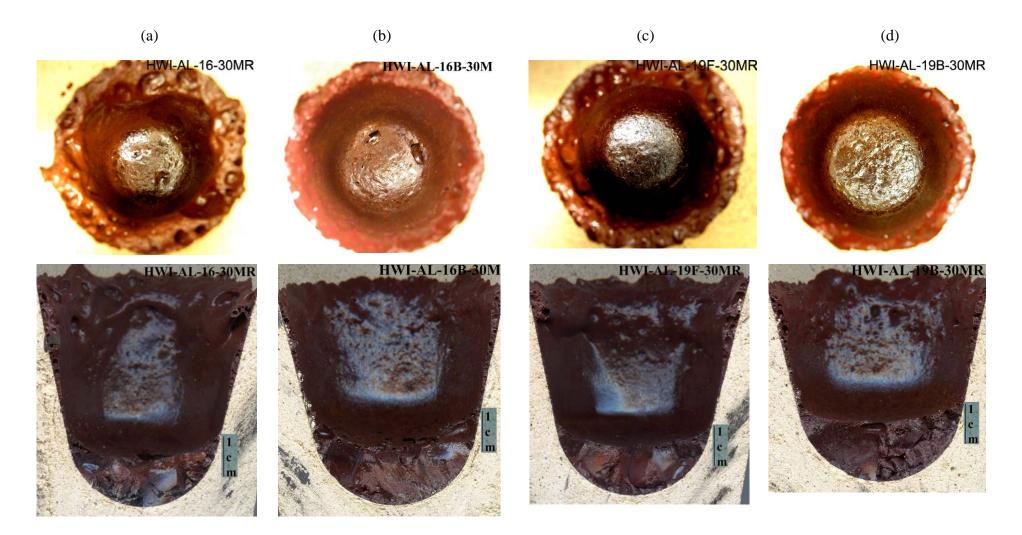
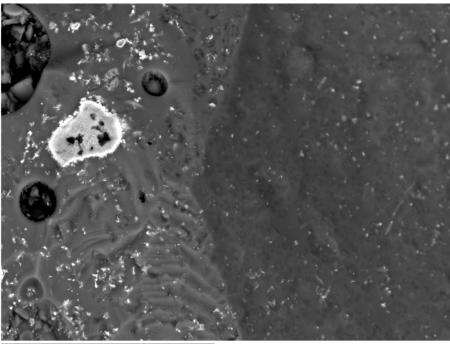
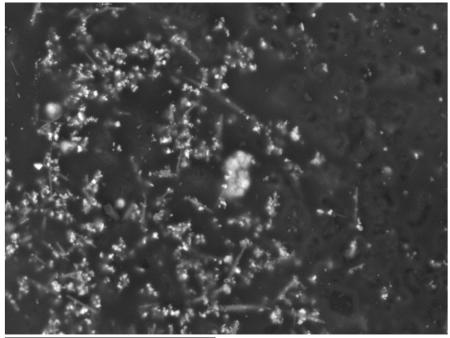


Figure 3.11. Top views and cross sections of feed samples after 30 minute test in VGF. (a) Feed F-HWI-Al-16 (Al<sub>2</sub>O<sub>3</sub>), (b) Feed F-HWI-Al-16B (Al(OH)<sub>3</sub>), (c) Feed F-HWI-Al-19F (Al<sub>2</sub>O<sub>3</sub>), and (d) Feed F-HWI-Al-19B (Al(OH)<sub>3</sub>).



60µm

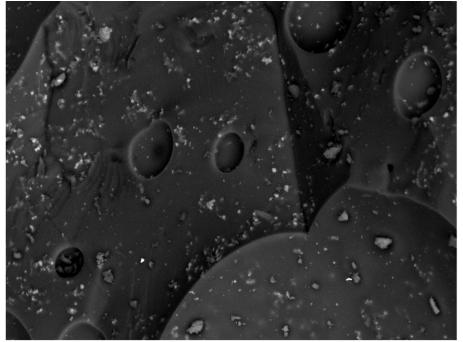
Electron Image 1



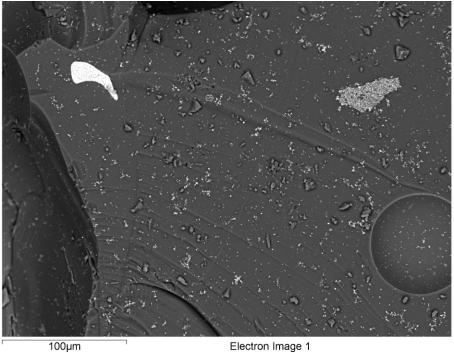
30µm

Electron Image 1

(a) Baseline feed for formulation HLW-E-Al-27 (Al<sub>2</sub>O<sub>3</sub>) Figure 3.12. SEM images of partially reacted feed after 30 minute VGF tests.

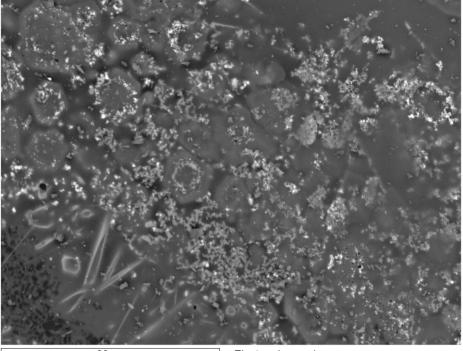


60μm Electron Image 1 (b) Feed for formulation HWI-Al-16 (Al<sub>2</sub>O<sub>3</sub>)

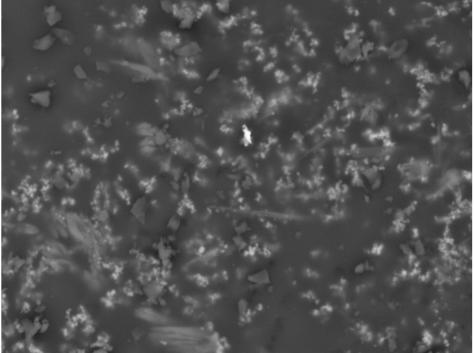


Liectron Image 1 (c) Feed for formulation HWI-Al-16 (Al(OH)<sub>3</sub>)

Figure 3.12. SEM images of partially reacted feed after 30 minute VGF tests (continued).



<sup>60μm</sup> Electron Image 1 (d) Feed for formulation HWI-Al-19 (Al<sub>2</sub>O<sub>3</sub>)



20µm Electron Image 1 (e) Feed for formulation HWI-Al-19 (Al(OH)<sub>3</sub>) Figure 3.12. SEM images of partially reacted feed after 30 minute VGF tests (continued).

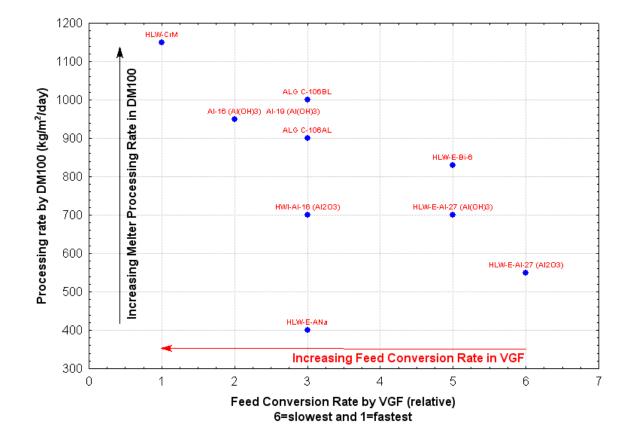


Figure 3.13. Comparison of melt rate ranking by VGF to the glass production rate determined by DM100 melter test. The definition of the VGF melt rate estimation is given in Table 3.5.

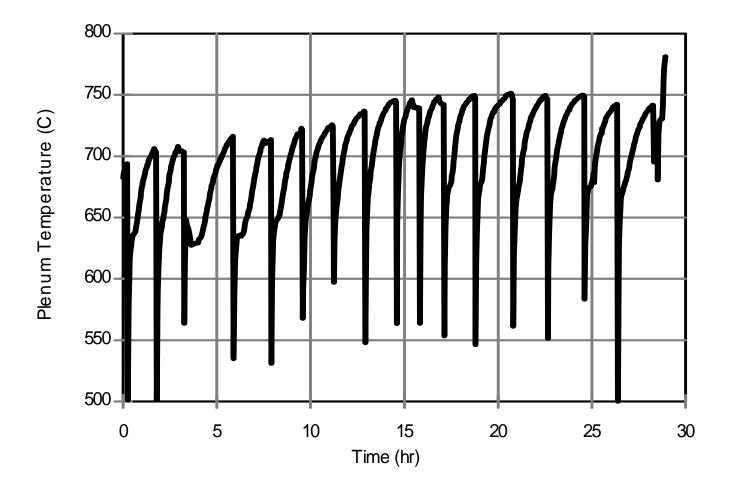


Figure 3.14. Changes in average plenum temperature over a series of DM10 feed consumption tests.

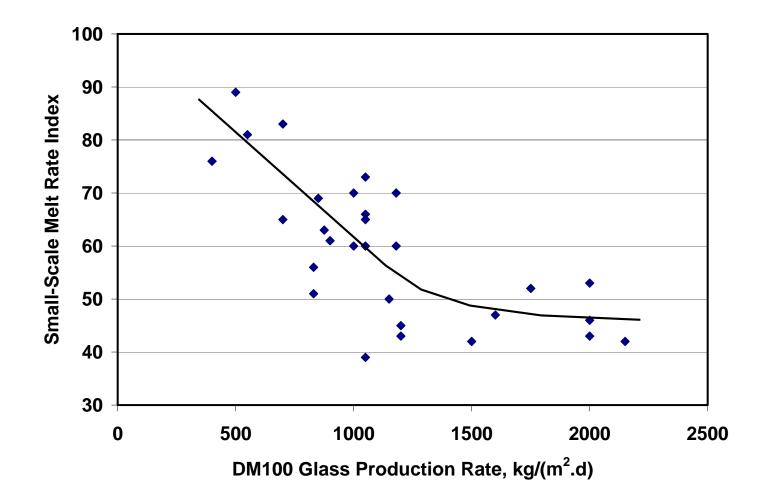


Figure 3.15. Correlation of DFC melt rate screening test data and DM100 production rates for a wide variety of feed compositions.

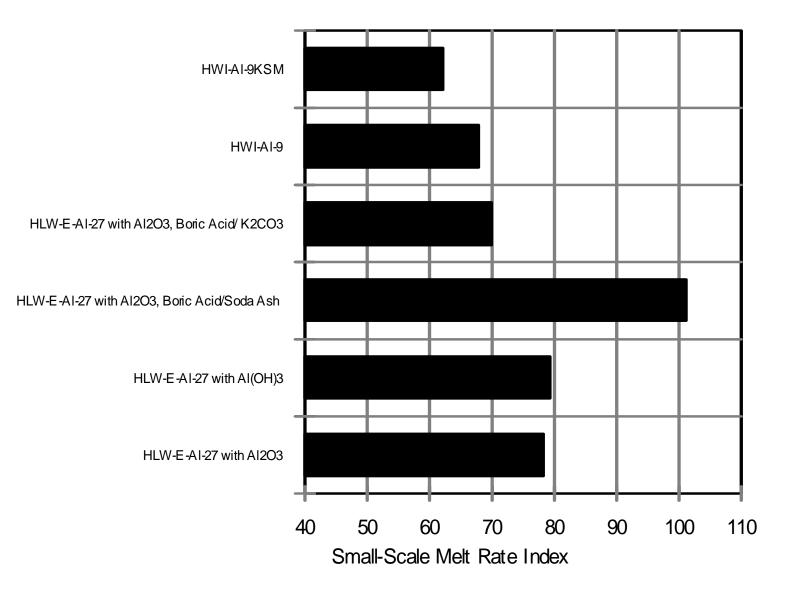


Figure 3.16. Comparison of base feeds from DFCmelt rate screening tests.

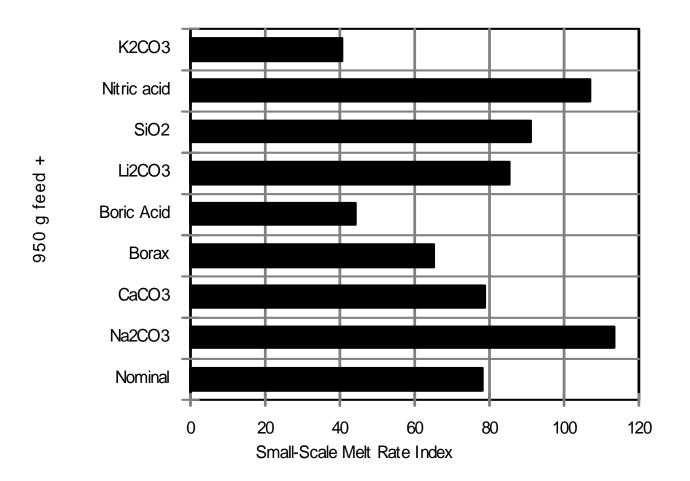


Figure 3.17. Comparison of additives combined with HLW-E-AL-27 (Al<sub>2</sub>O<sub>3</sub>) from DFC melt rate screening tests.

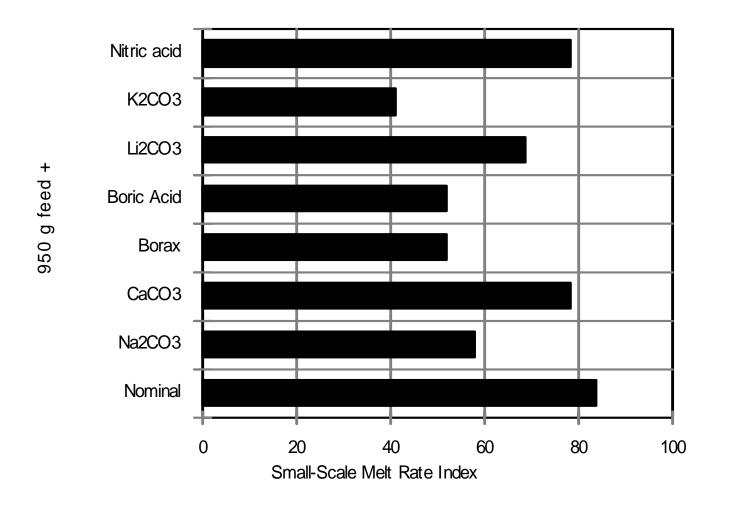


Figure 3.18. Comparison of additives combined with HLW-E-AL-27 (Al(OH)<sub>3</sub>) from DFC melt rate screening tests.

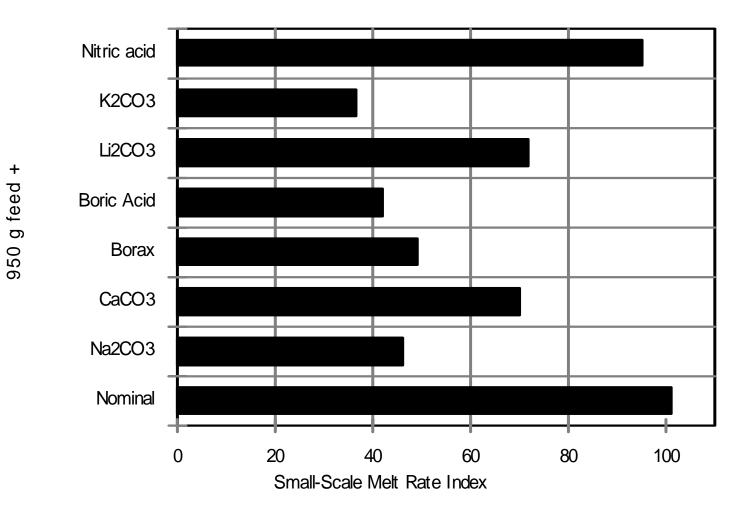


Figure 3.19. Comparison of additives combined with HLW-E-AL-27 (Al<sub>2</sub>O<sub>3</sub>+boric acid/soda ash replacing borax) from DFC melt rate screening tests.

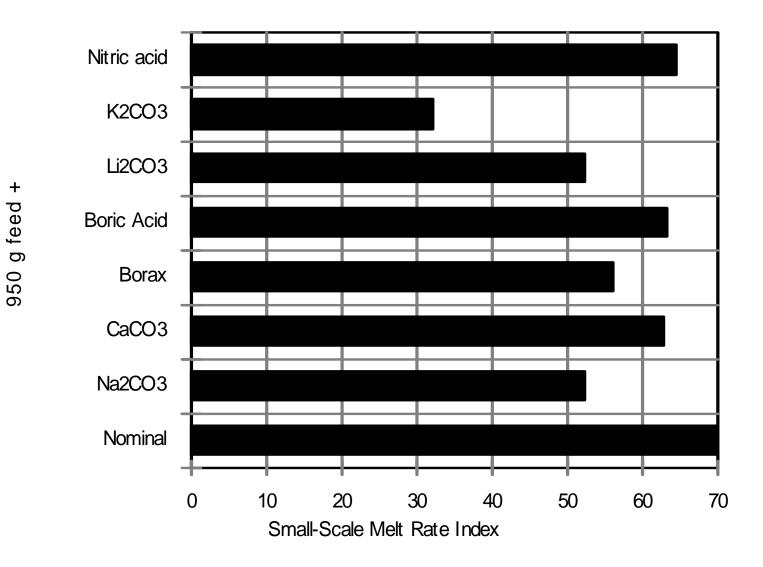


Figure 3.20. Comparison of additives combined with HLW-E-AL-27 (Al<sub>2</sub>O<sub>3</sub>+boric acid/potassium carbonate replacing borax) from DFC melt rate screening tests.

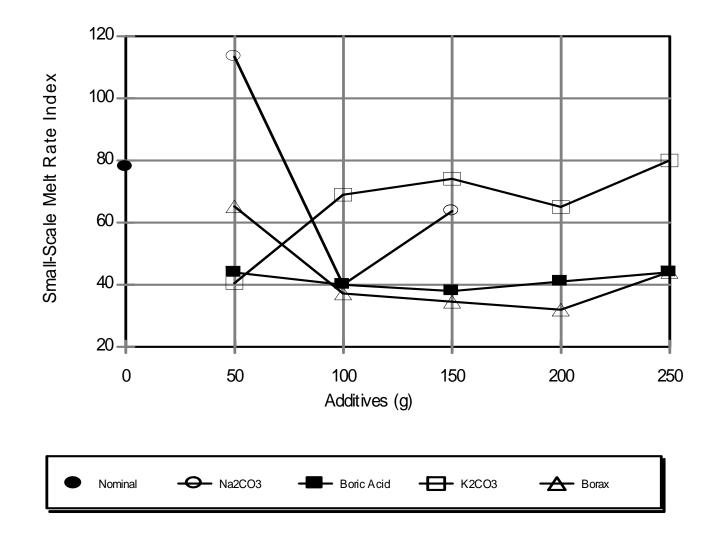


Figure 3.21. Comparison of additives and amounts combined with HLW-E-AL-27 (Al<sub>2</sub>O<sub>3</sub>) from DFC melt rate screening tests.

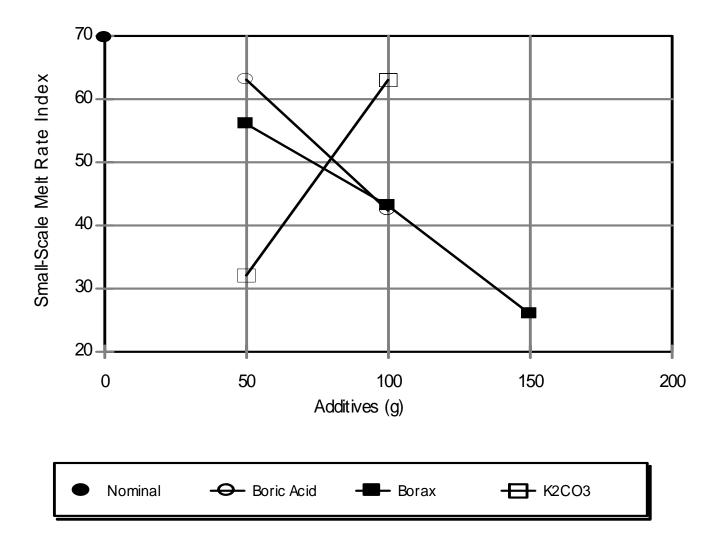


Figure 3.22. Comparison of additives and amounts combined with HLW-E-AL-27 (Al<sub>2</sub>O<sub>3</sub>+boric acid/potassium carbonate replacing borax) from DFC melt rate screening tests.

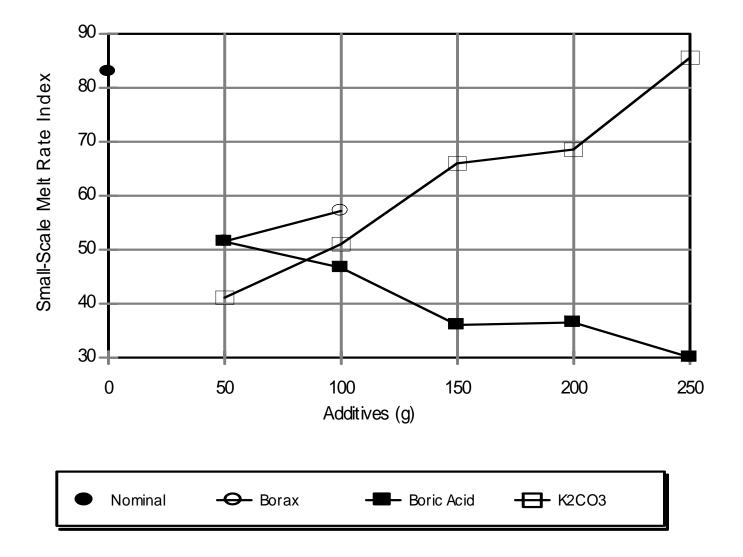


Figure 3.23. Comparison of additives and amounts combined with HLW-E-AL-27 (Al(OH)<sub>3</sub>) from DFC melt rate screening tests.

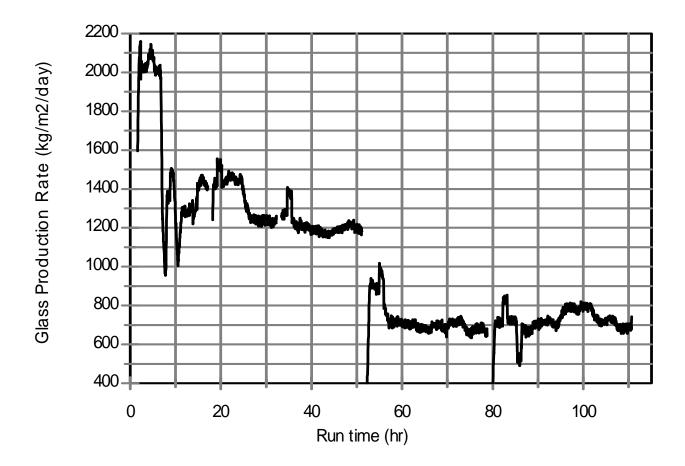


Figure 4.1.a. Glass production rates (hourly moving averages) for DM100 Tests 1 and 2.

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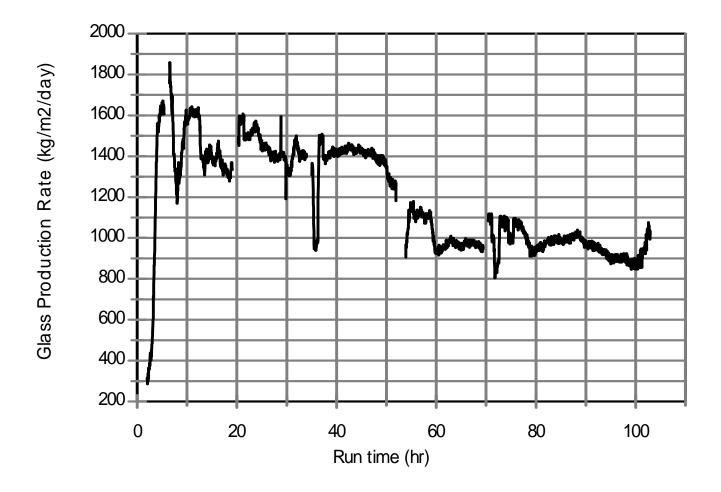


Figure 4.1.b. Glass production rates (hourly moving averages) for DM100 Tests 3 and 4.

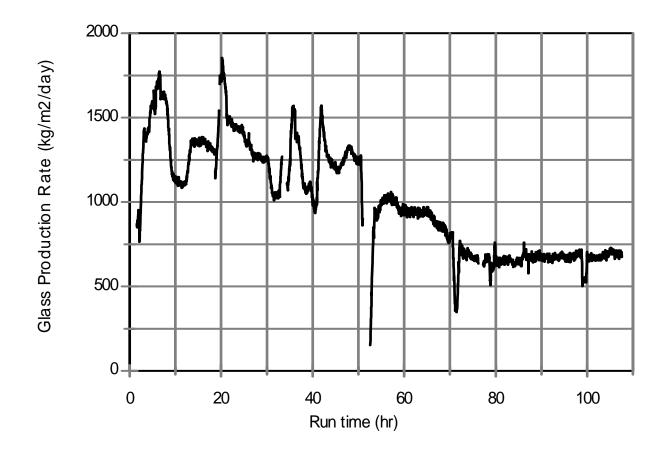


Figure 4.1.c. Glass production rates (hourly moving averages) for DM100 Tests 5 and 6.

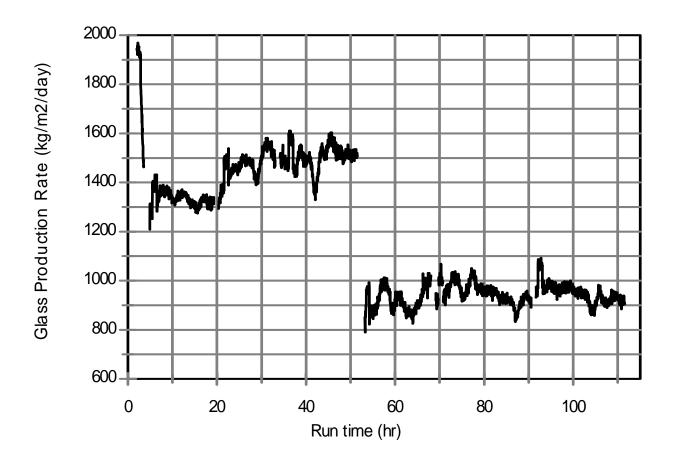


Figure 4.1.d. Glass production rates (hourly moving averages) for DM100 Tests 7 and 8.

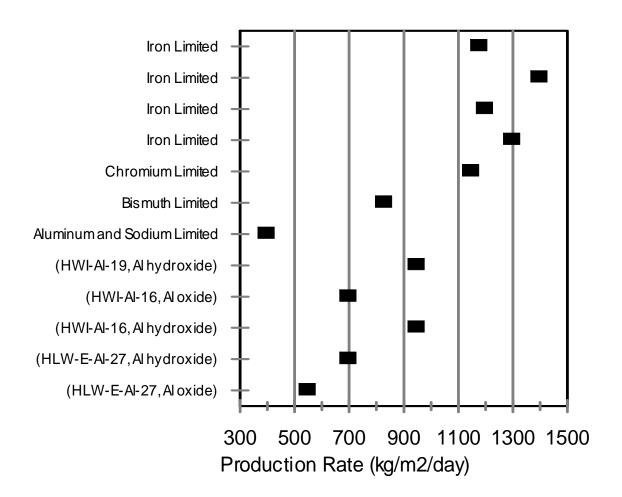


Figure 4.2. Steady-state glass production rates during DM100 tests at constant bubbling rate, glass temperature (1150°C), and feed solids content 500 (±50) g glass per liter feed.

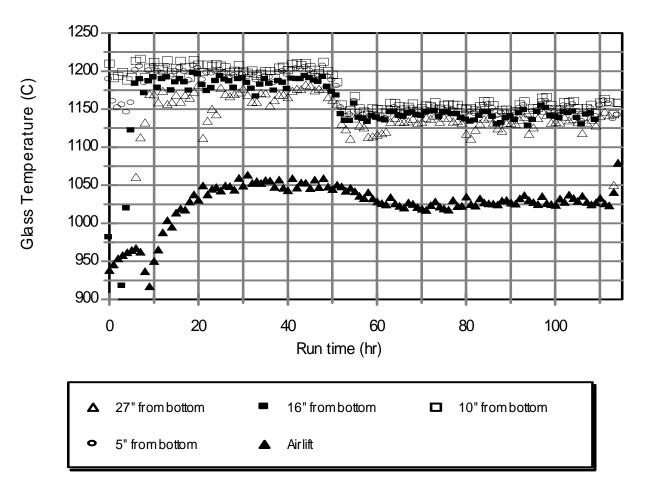


Figure 4.3.a. Glass temperatures (hourly averages) during DM100 Tests 1 and 2.

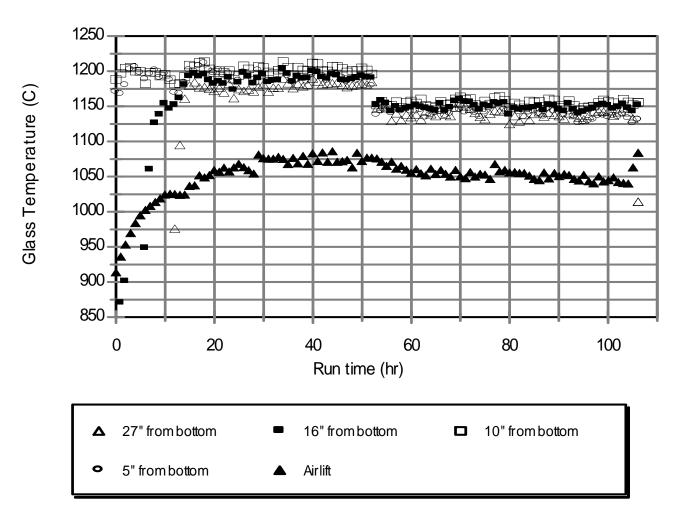


Figure 4.3.b. Glass temperatures (hourly averages) during DM100 Tests 3 and 4.

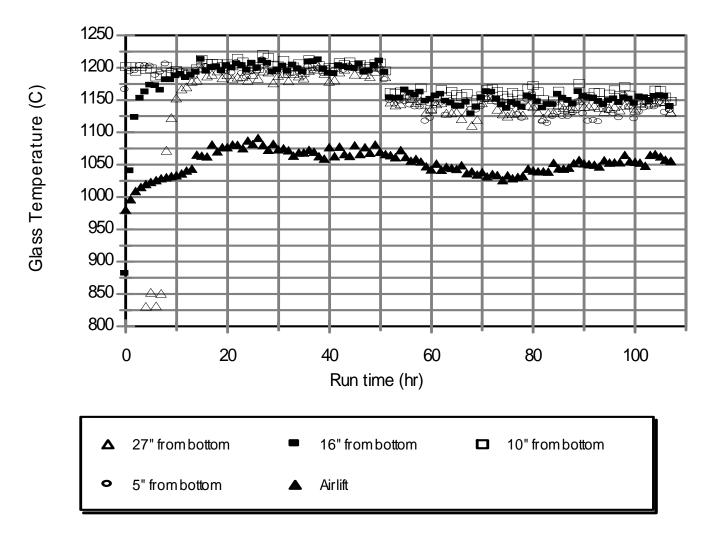


Figure 4.3.c. Glass temperatures (hourly averages) during DM100 Tests 5 and 6.

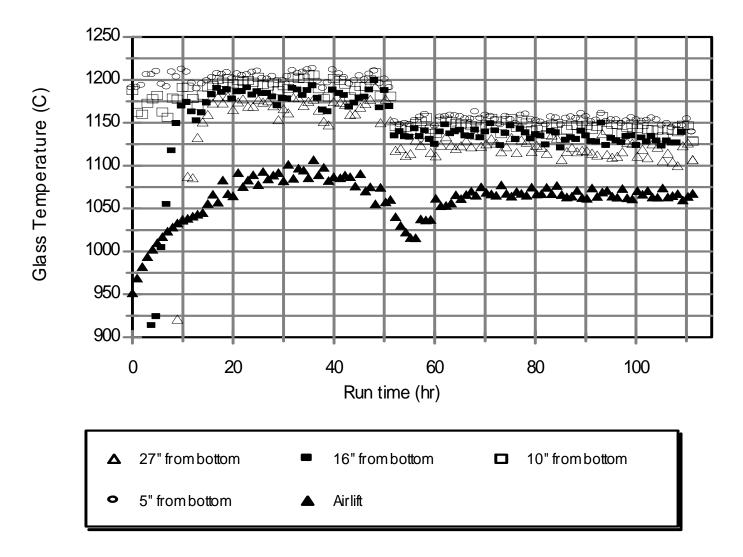


Figure 4.3.d. Glass temperatures (hourly averages) during DM100 Tests 7 and 8.

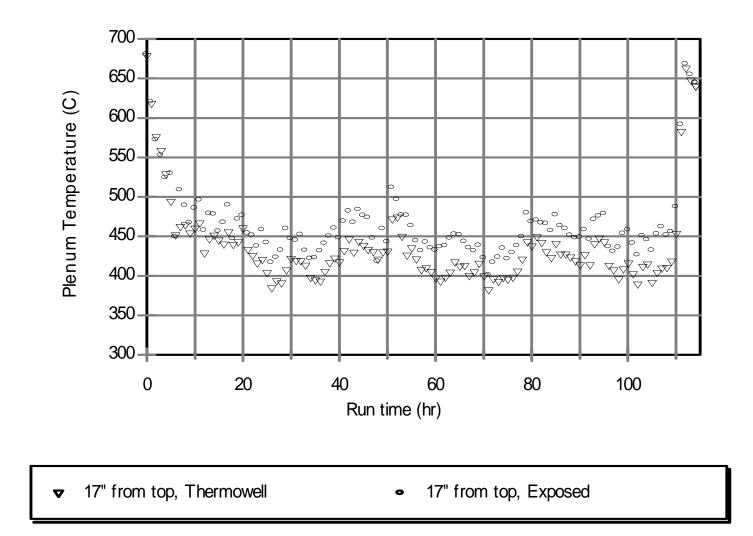


Figure 4.4.a. Plenum temperatures (hourly averages) during DM100 Tests 1 and 2.

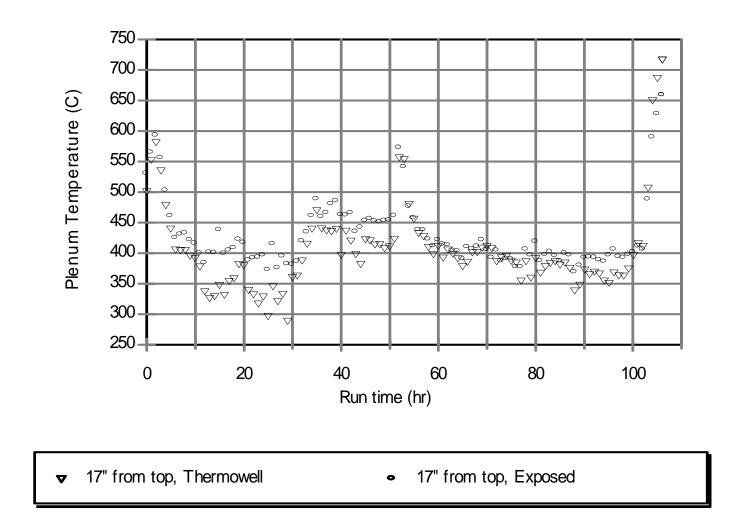


Figure 4.4.b. Plenum temperatures (hourly averages) during DM100 Tests 3 and 4.

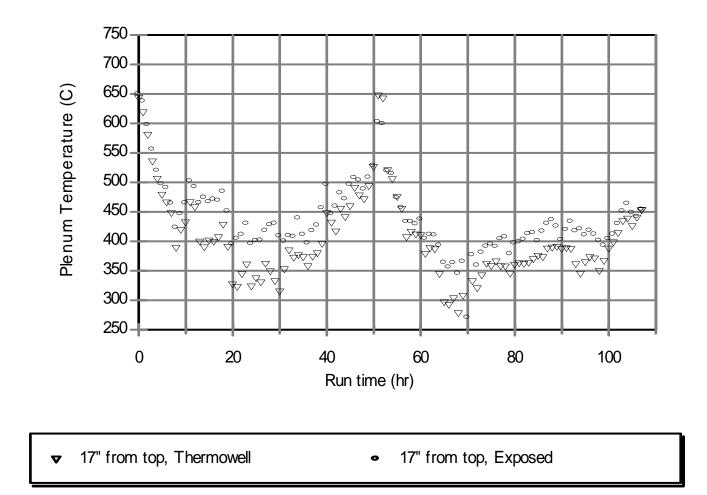


Figure 4.4.c. Plenum temperatures (hourly averages) during DM100 Tests 5 and 6.

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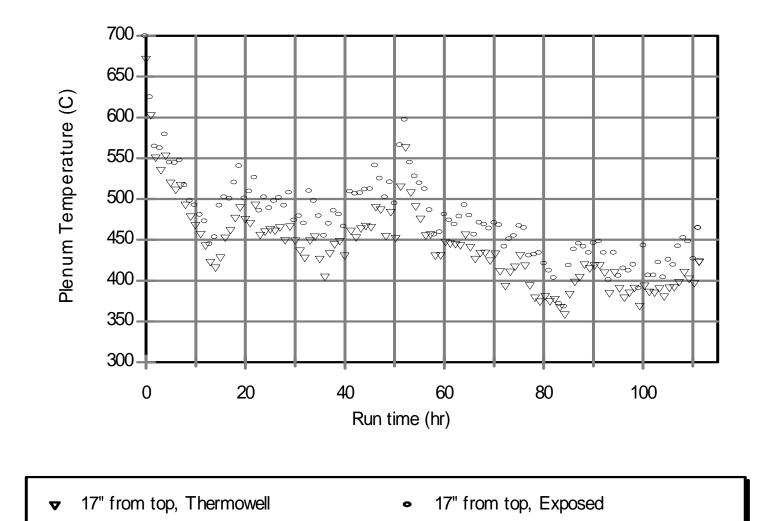


Figure 4.4.d. Plenum temperatures (hourly averages) during DM100 Tests 7 and 8.

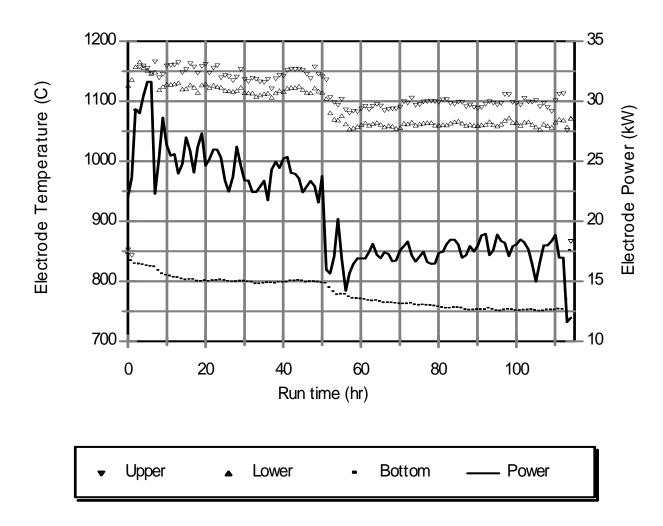


Figure 4.5.a. Electrode temperatures and power (hourly averages) during DM100 Tests 1 and 2.

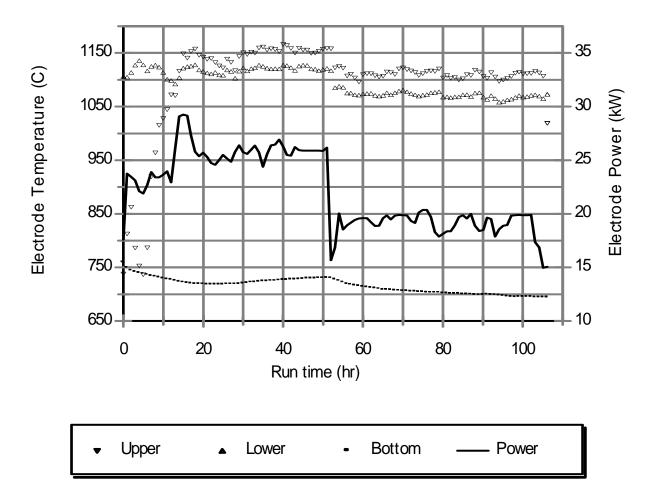


Figure 4.5.b. Electrode temperatures and power (hourly averages) during DM100 Tests 3 and 4.

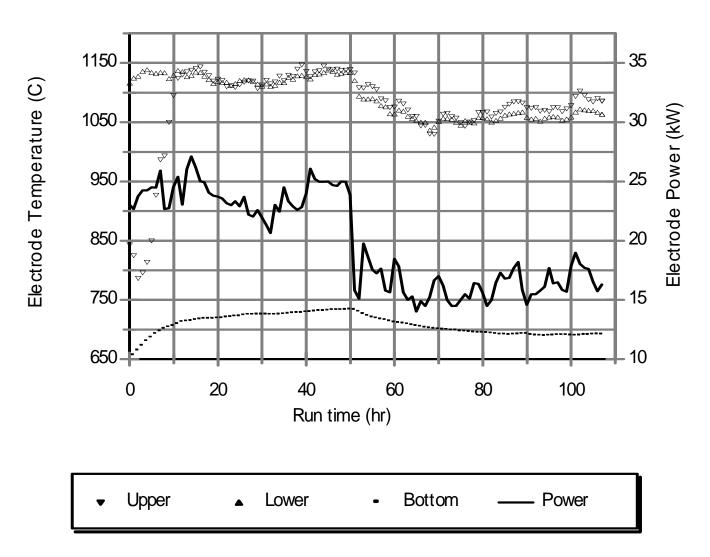


Figure 4.5.c. Electrode temperatures and power (hourly averages) during DM100 Tests 5 and 6.

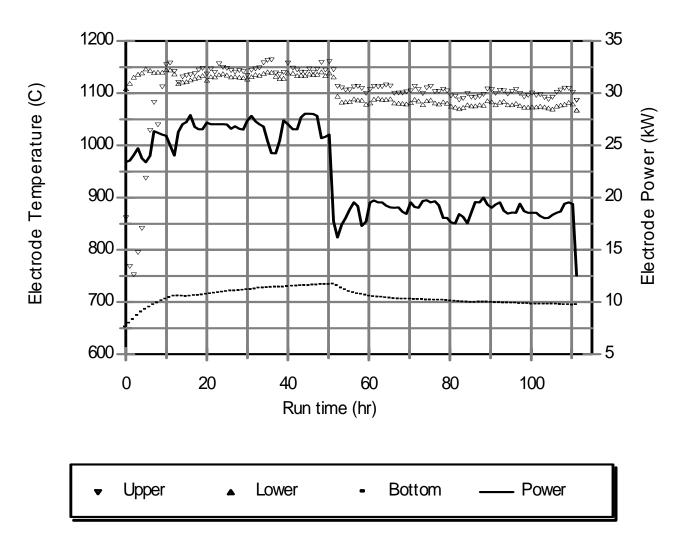


Figure 4.5.d. Electrode temperatures and power (hourly averages) during DM100 Tests 7 and 8.

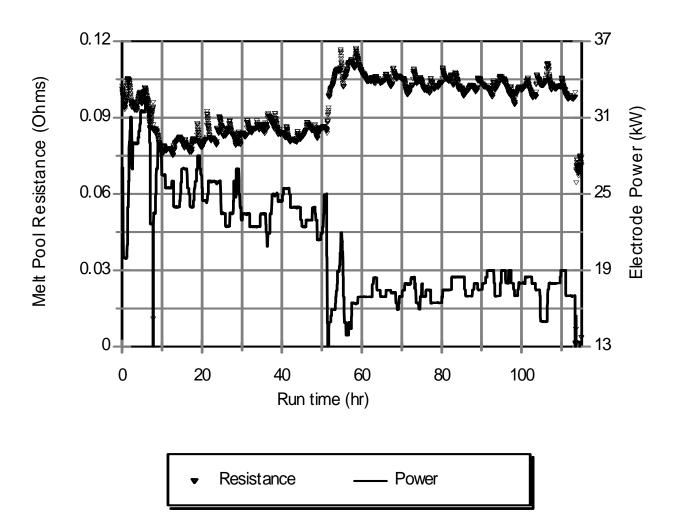


Figure 4.6.a. Melt pool resistance and total electrode power during DM100 Tests 1 and 2.

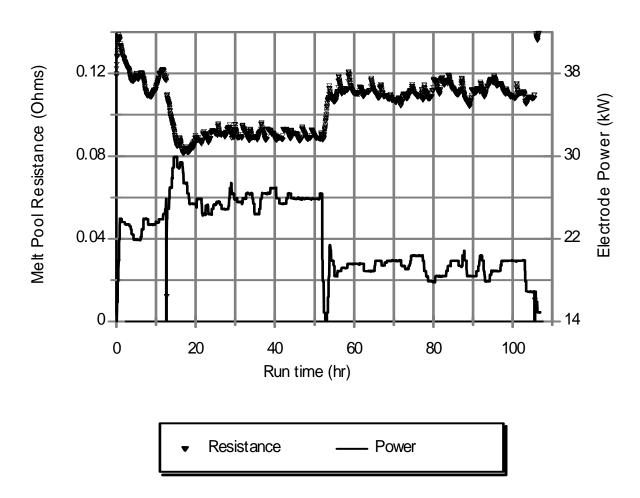


Figure 4.6.b. Melt pool resistance and total electrode power during DM100 Tests 3 and 4.

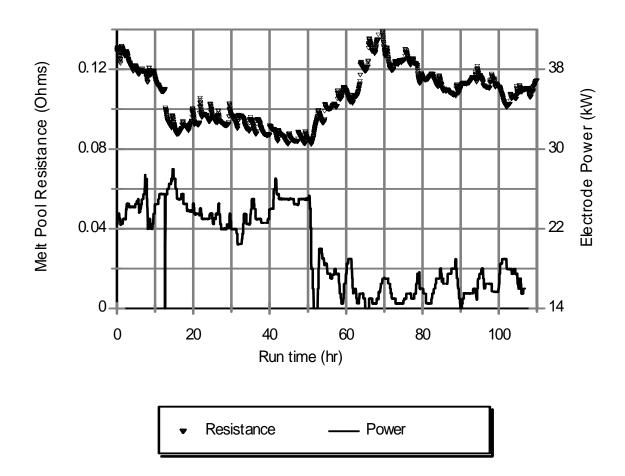


Figure 4.6.c. Melt pool resistance and total electrode power during DM100 Tests 5 and 6.

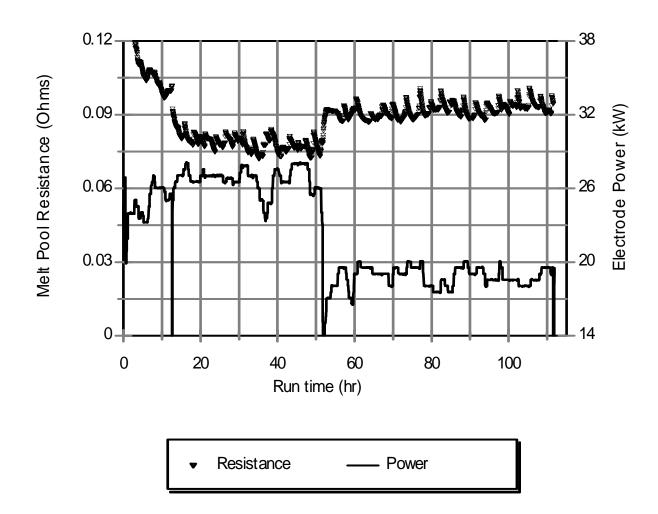


Figure 4.6.d. Melt pool resistance and total electrode power during DM100 Tests 7 and 8.

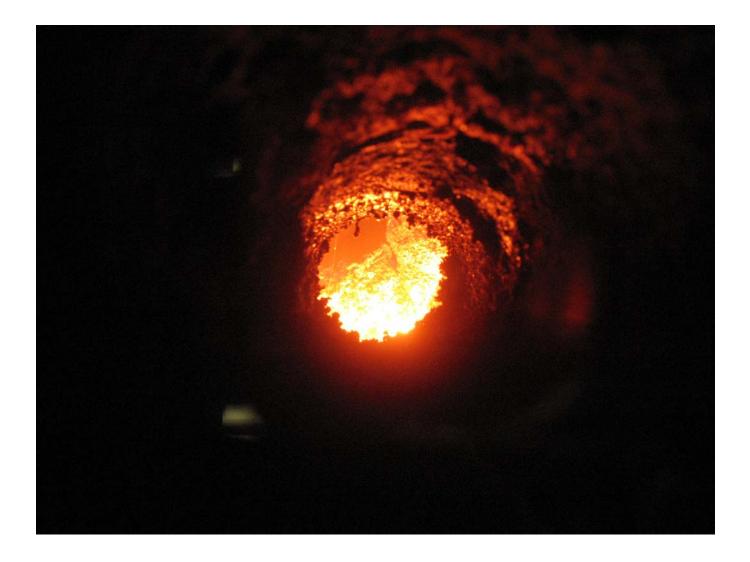


Figure 5.1. Picture of cold cap through north view port showing the thickness of the cold cap.

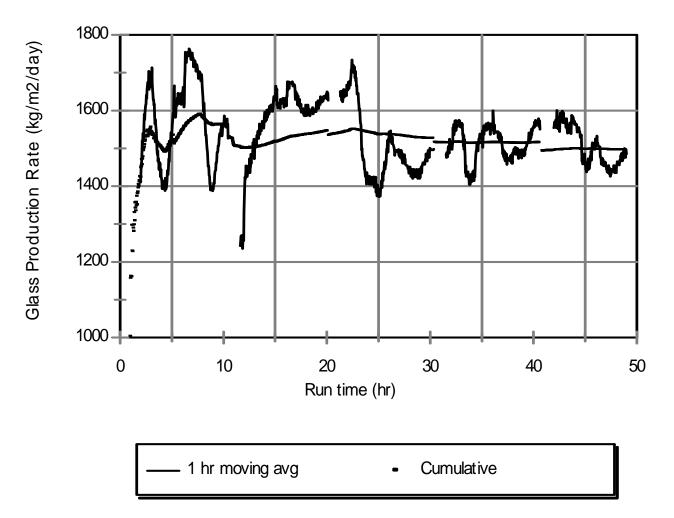


Figure 5.2.a. Production rates for DM1200 Test 1.

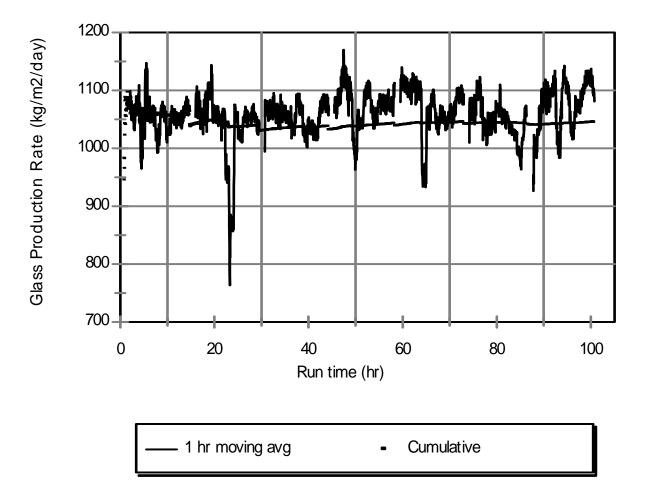


Figure 5.2.b. Production rates for DM1200 Tests 2 and 3.

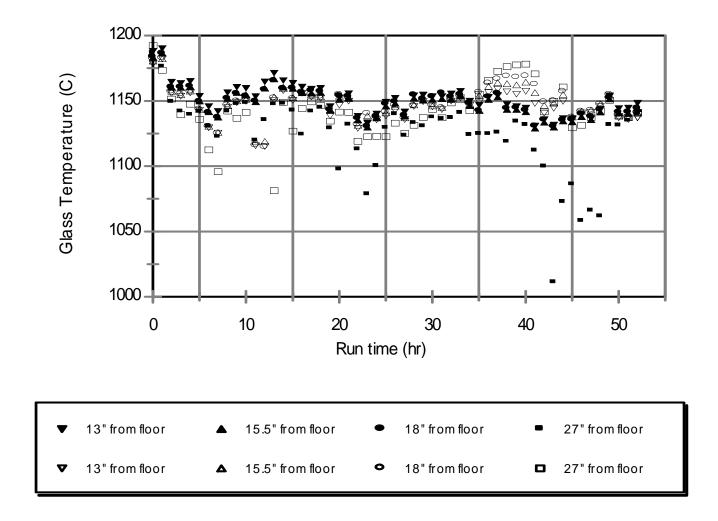


Figure 5.3.a. Glass temperatures (hourly averages) for DM1200 Test 1.

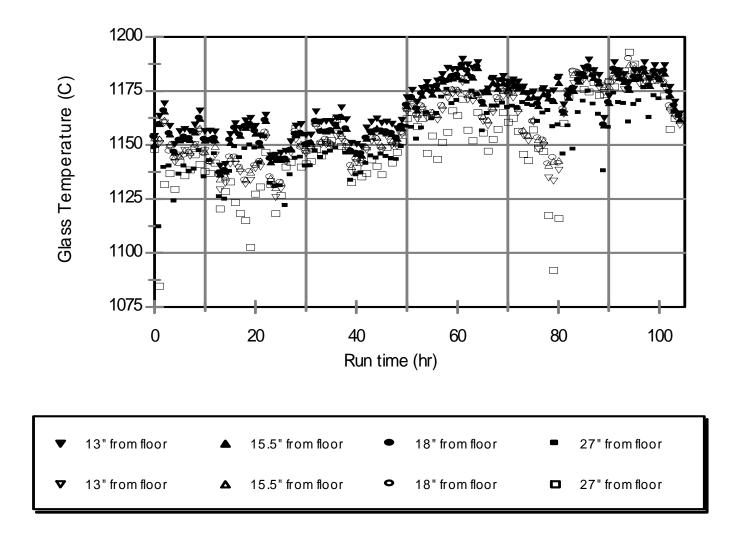


Figure 5.3.b. Glass temperatures (hourly averages) for DM1200 Tests 2 and 3.

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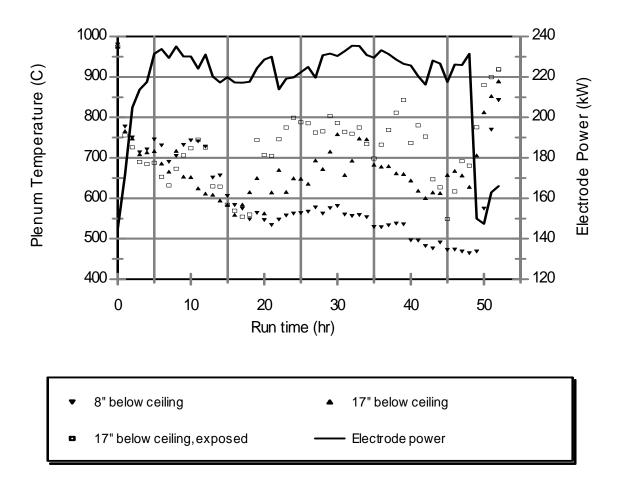


Figure 5.4.a. Plenum temperatures and electrode power (hourly averages) for DM1200 Test 1.

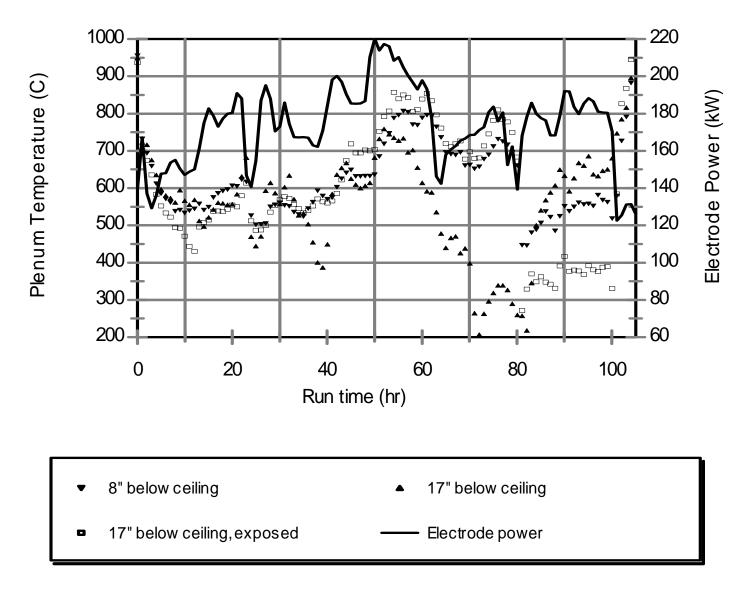


Figure 5.4.b. Plenum temperatures and electrode power (hourly averages) for DM1200 Tests 2 and 3.

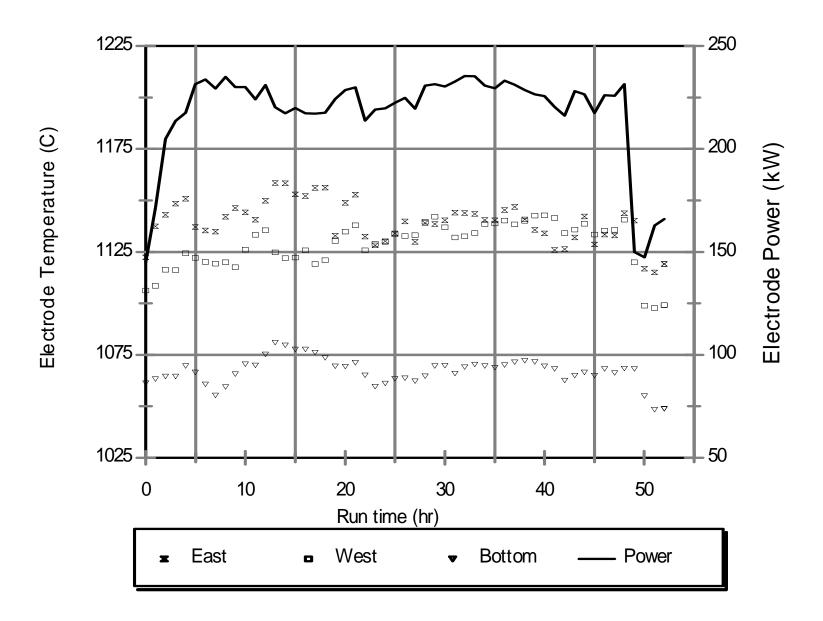


Figure 5.5.a. Electrode temperatures and power (hourly averages) for DM1200 Test 1.

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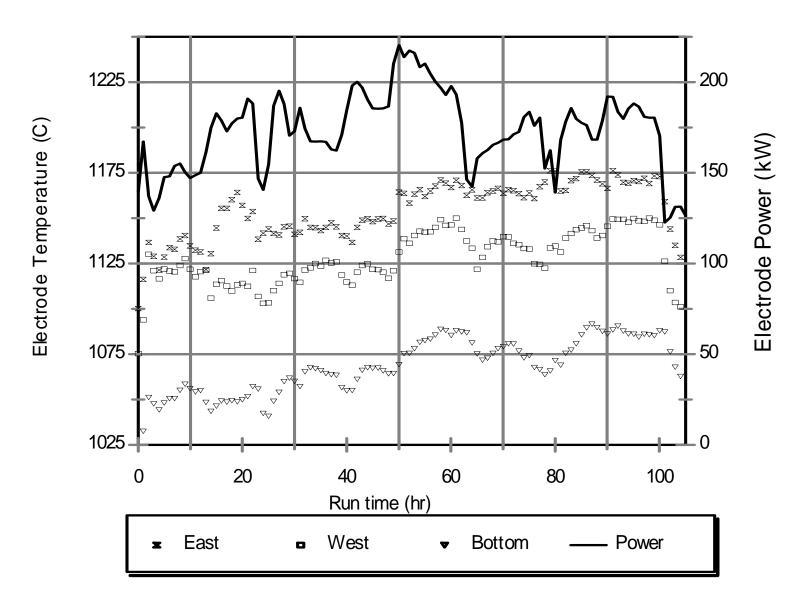


Figure 5.5.b. Electrode temperatures and power (hourly averages) for DM1200 Tests 2 and 3.



Figure 5.6.a. Electrode power and glass resistance for DM1200 Test 1.

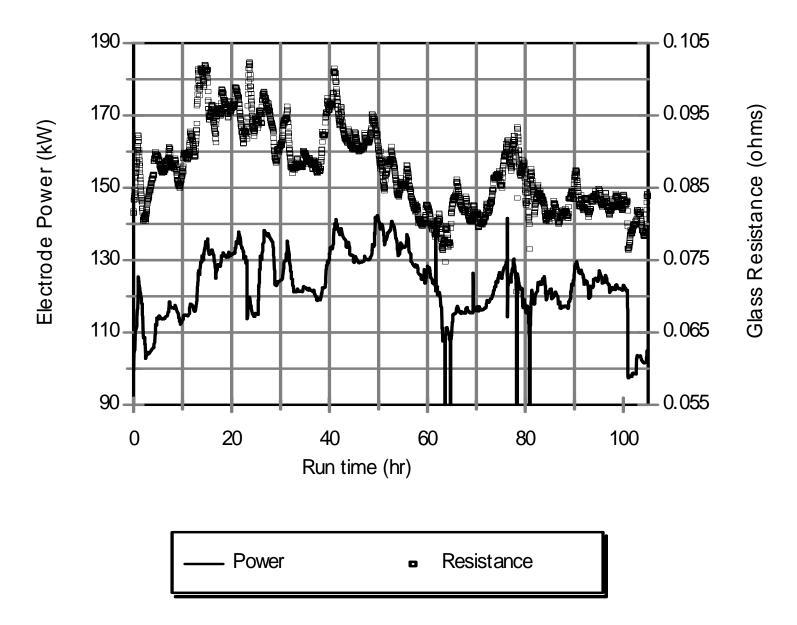


Figure 5.6.b. Electrode power and glass resistance for DM1200 Tests 2 and 3.

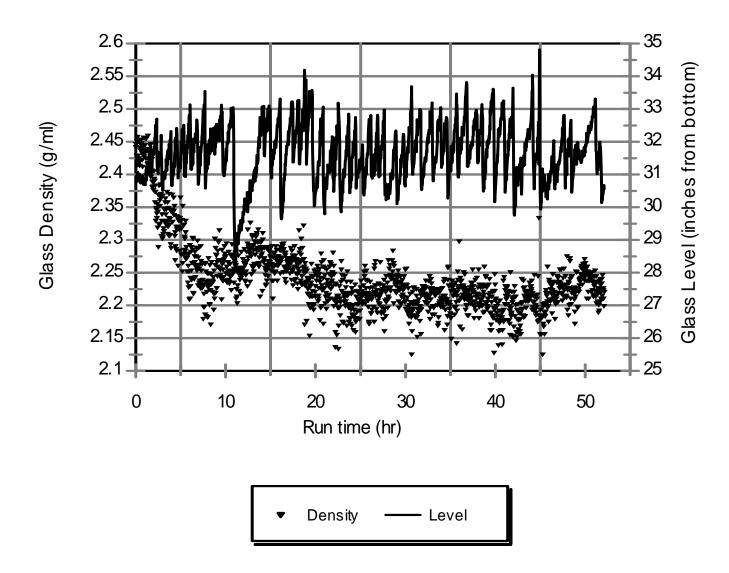


Figure 5.7.a. Glass density and level for DM1200 Test 1.

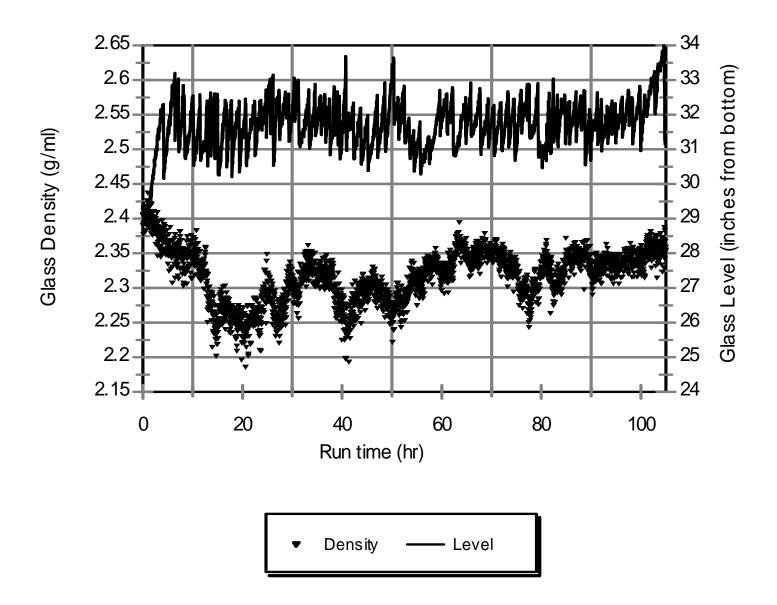


Figure 5.7.b. Glass density and level for DM1200 Tests 2 and 3.

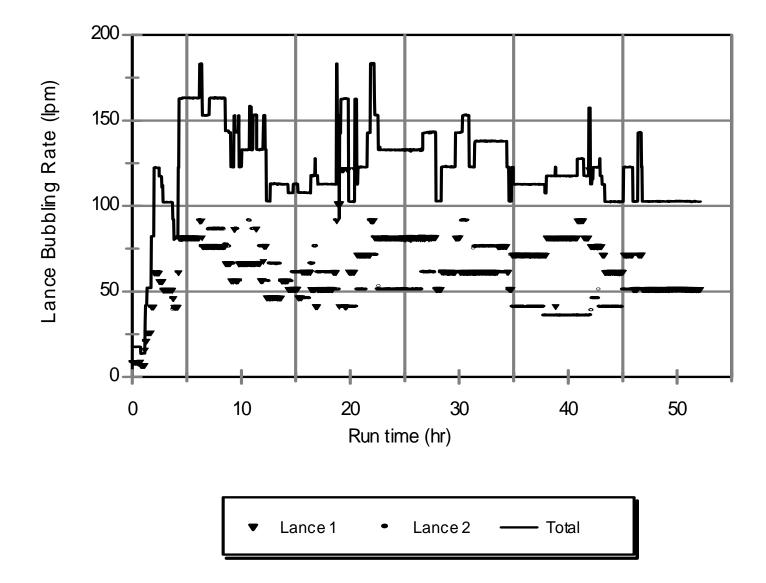


Figure 5.8.a. Glass pool bubbling for DM1200 Test 1.

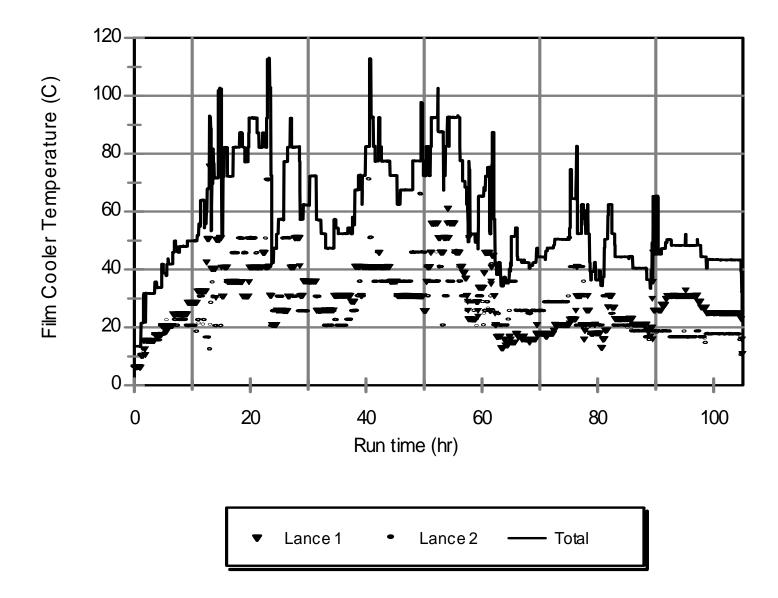


Figure 5.8.b. Glass pool bubbling for DM1200 Tests 2 and 3.

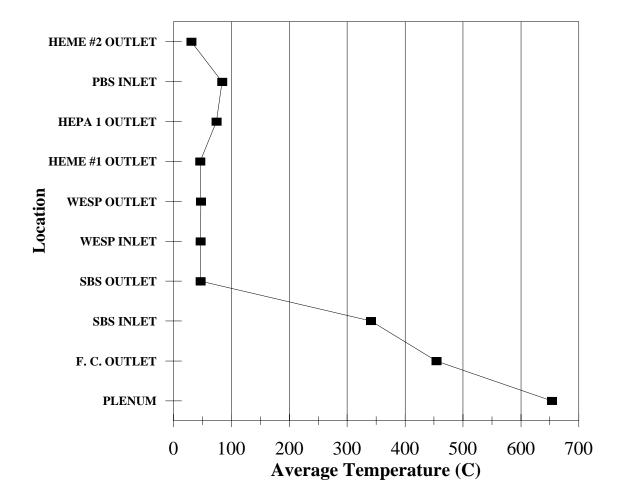


Figure 5.9.a. Average gas temperatures along the DM1200 off-gas train during Test 1.

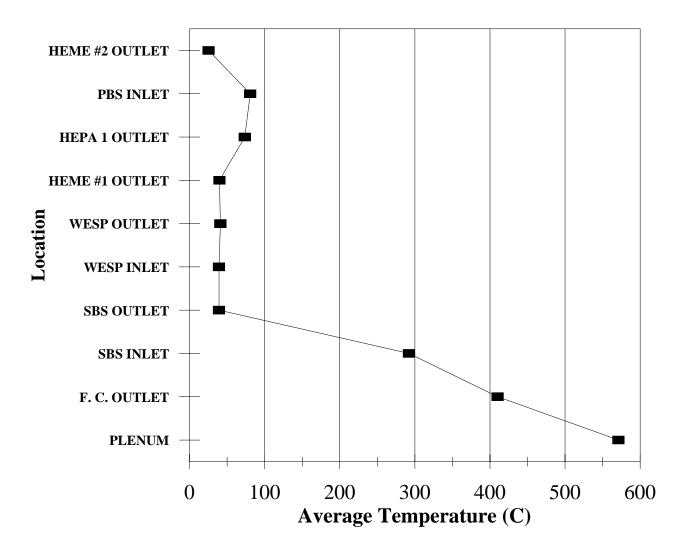


Figure 5.9.b. Average gas temperatures along the DM1200 off-gas train during Test 2.

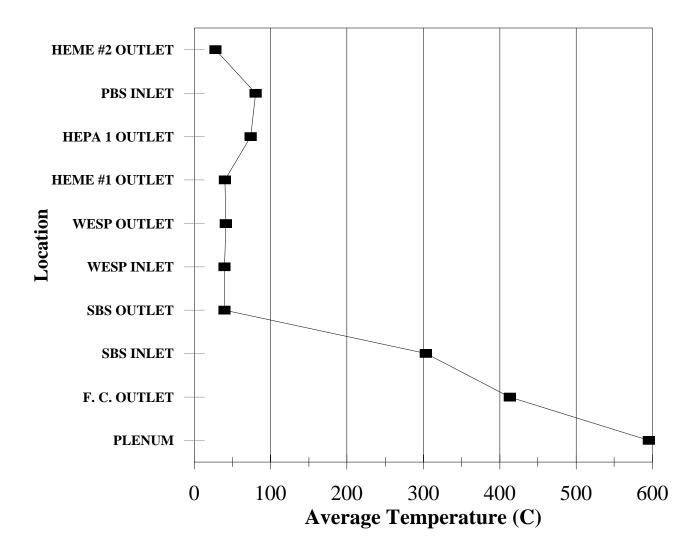


Figure 5.9.c. Average gas temperatures along the DM1200 off-gas train during Test 3.

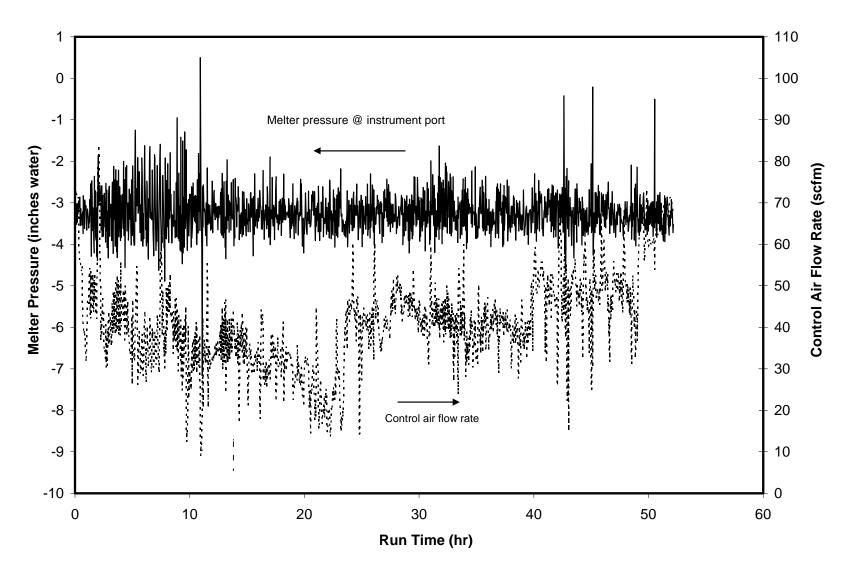


Figure 5.10.a. Melter pressure at instrument port and control air flow rate during Test 1.

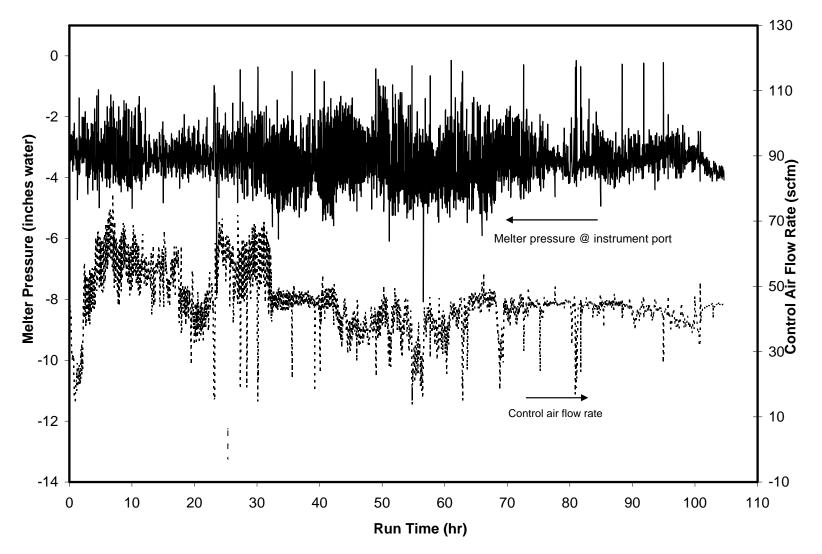


Figure 5.10.b. Melter pressure at instrument port and control air flow rate during Tests 2 and 3.

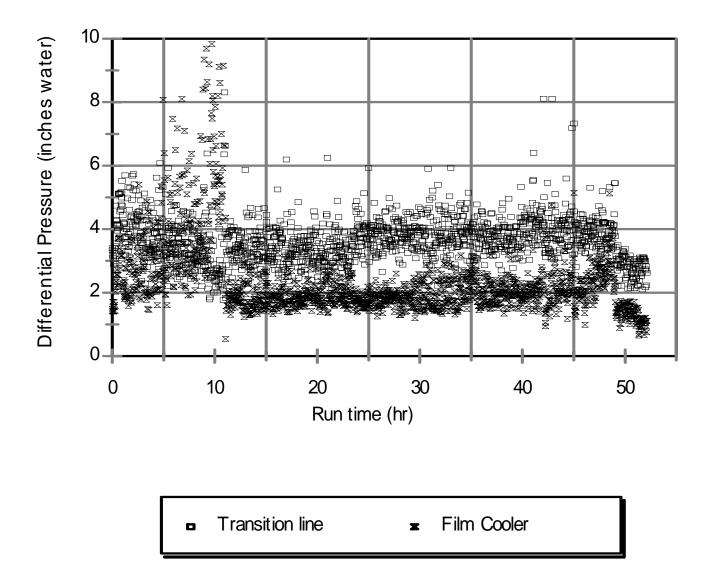


Figure 5.11.a. Differential pressure across the transition line and film cooler during DM1200 Test 1.

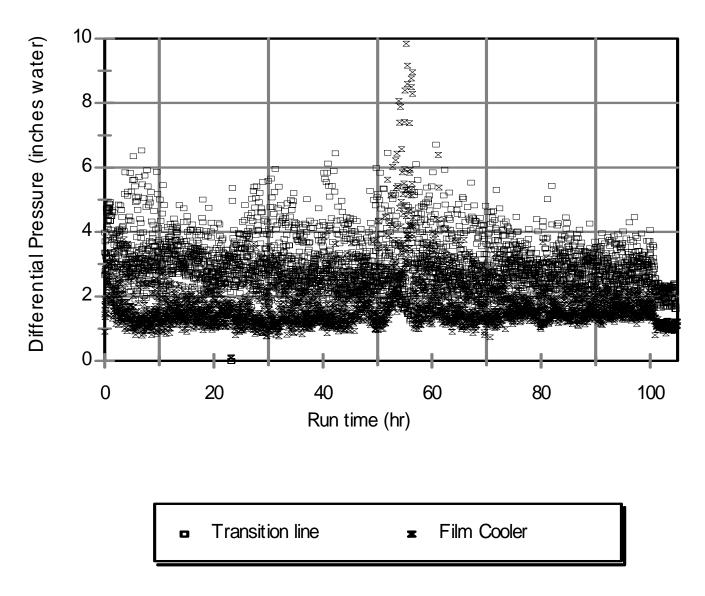


Figure 5.11.b. Differential pressure across the transition line and film cooler during DM1200 Tests 2 and 3.

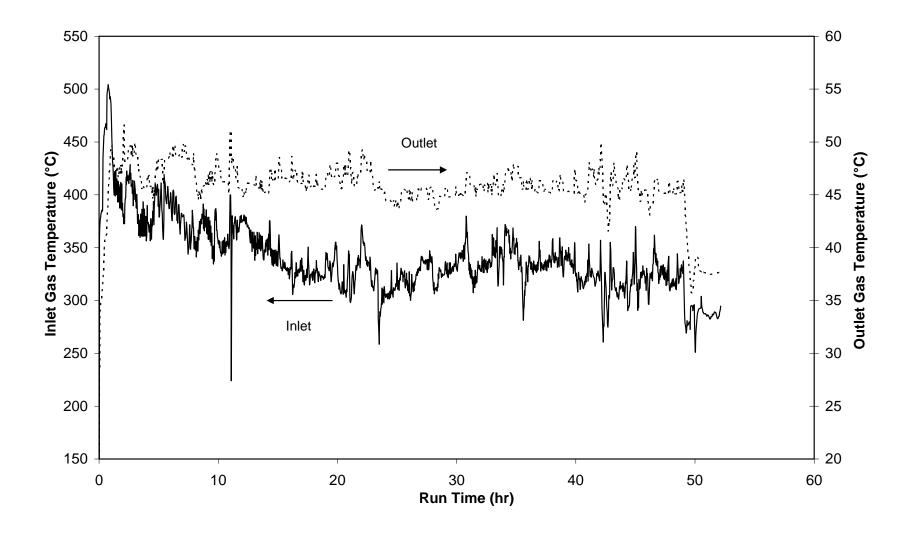


Figure 5.12.a. SBS inlet and outlet gas temperatures during Test 1.

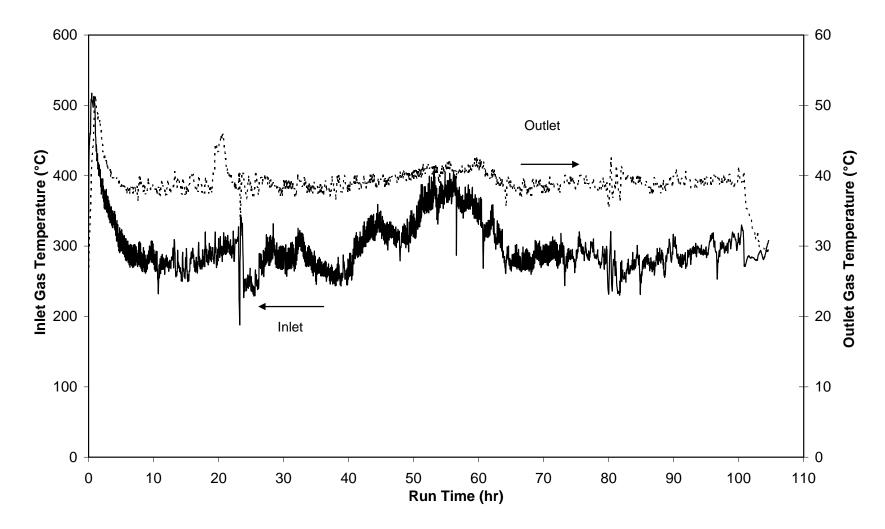


Figure 5.12.b. SBS inlet and outlet gas temperatures during Tests 2 and 3.

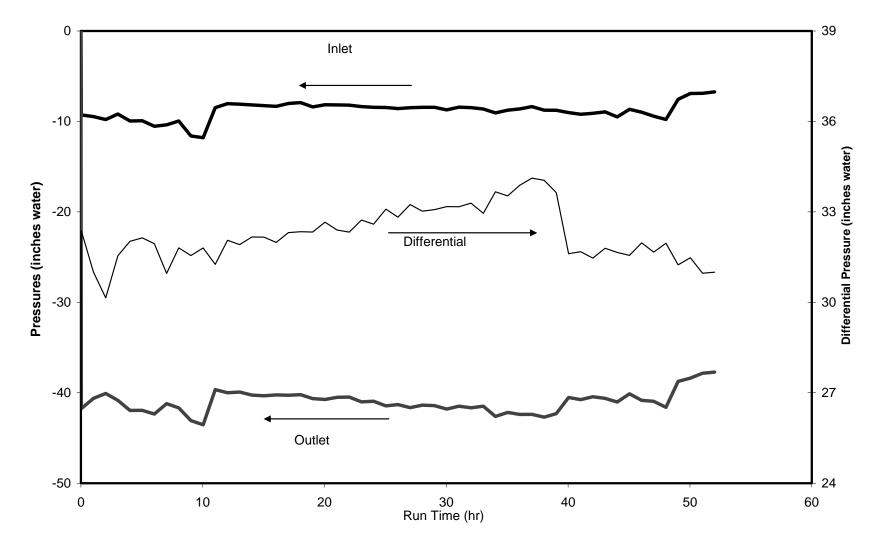


Figure 5.13.a. SBS inlet, outlet, and differential pressures (hourly average values) during Test 1.

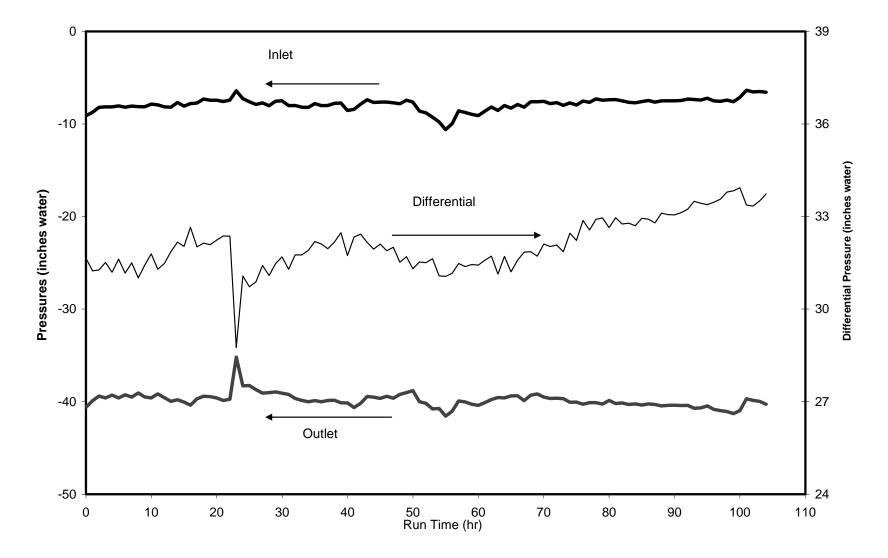


Figure 5.13.b. SBS inlet, outlet, and differential pressures (hourly average values) during Tests 2 and 3.

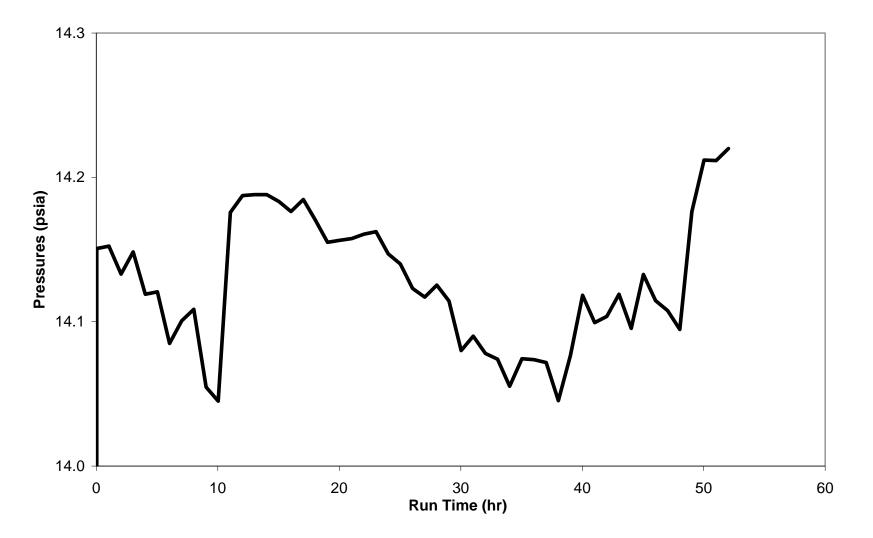


Figure 5.14.a. SBS downcomer annulus pressure (hourly average values) during Test 1.

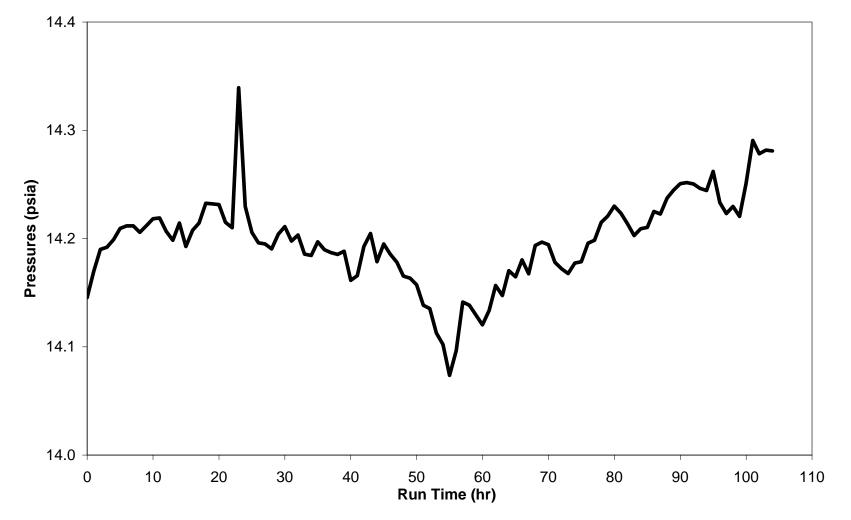


Figure 5.14.b. SBS downcomer annulus pressure (hourly average values) during Tests 2 and 3.

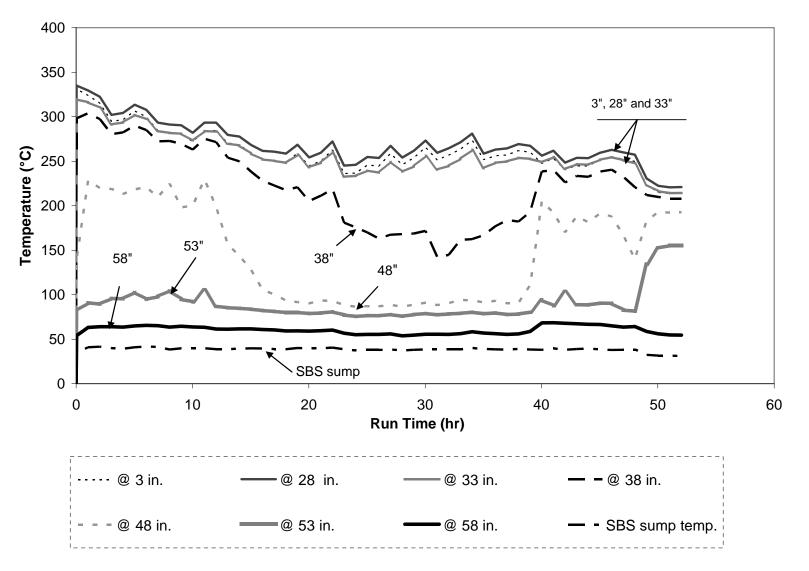
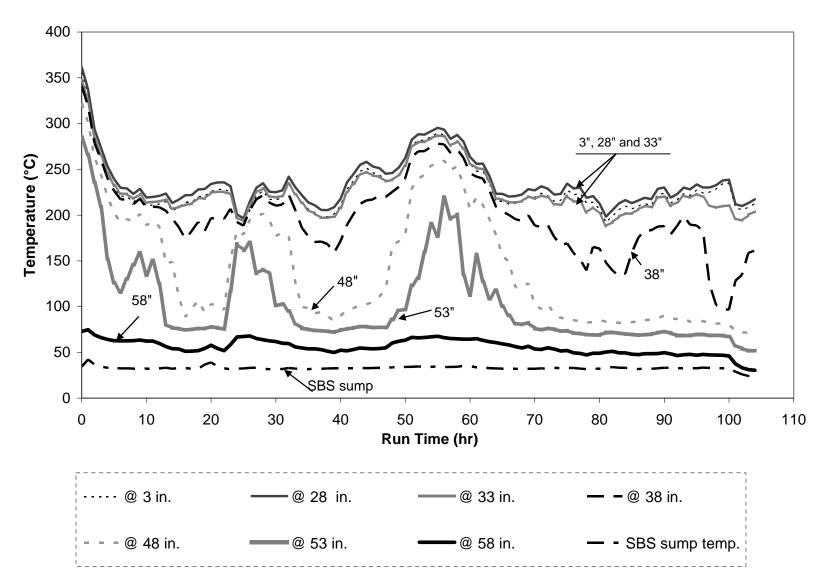


Figure 5.15.a. Off-gas temperatures in the SBS downcomer and sump water temperatures (hourly average values) during Test 1.





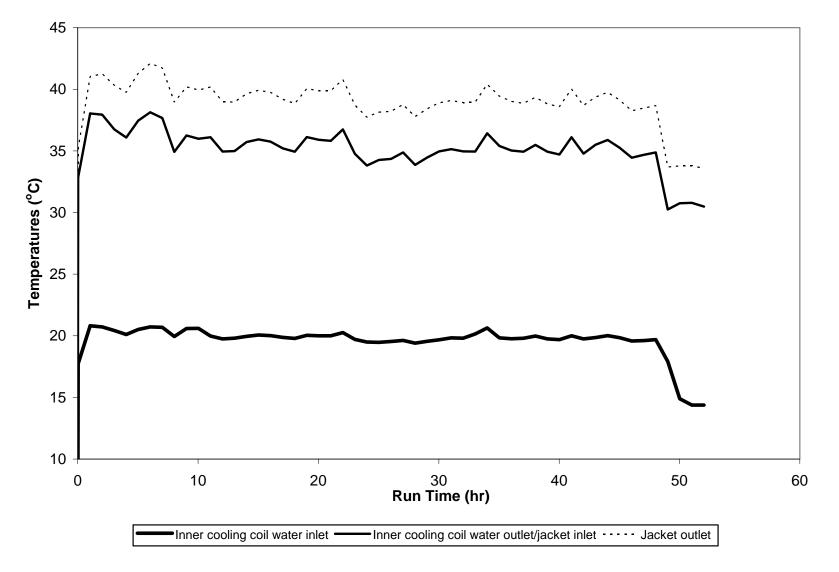


Figure 5.16.a. SBS cooling coil inlet, cooling coil outlet/jacket inlet and jacket outlet water temperatures (hourly average values) during Test 1.

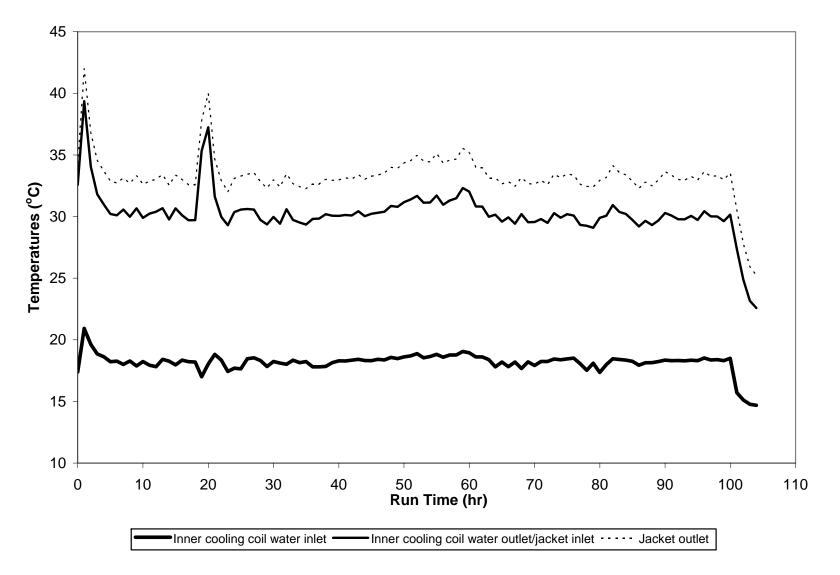


Figure 5.16.b. SBS cooling coil inlet, cooling coil outlet/jacket inlet and jacket outlet water temperatures (hourly average values) during Tests 2 and 3.

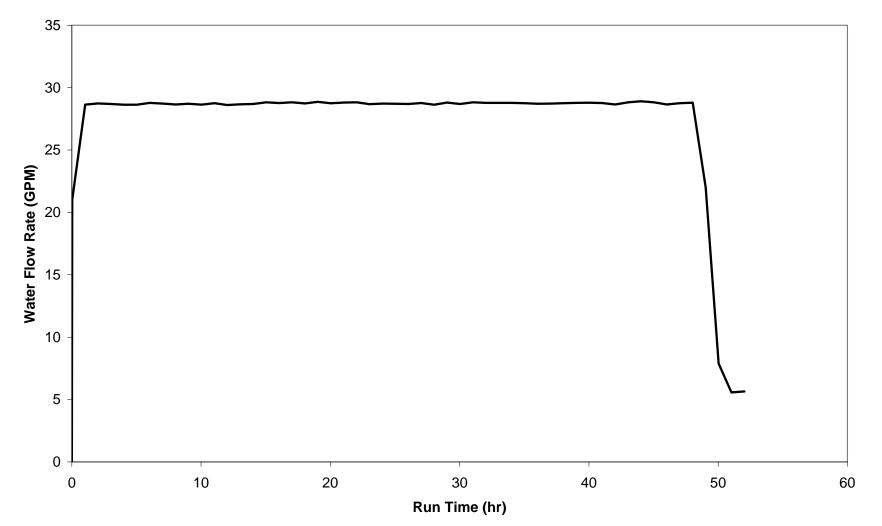


Figure 5.17.a. SBS cooling coil/jacket water flow rate (hourly average values) during Test 1.

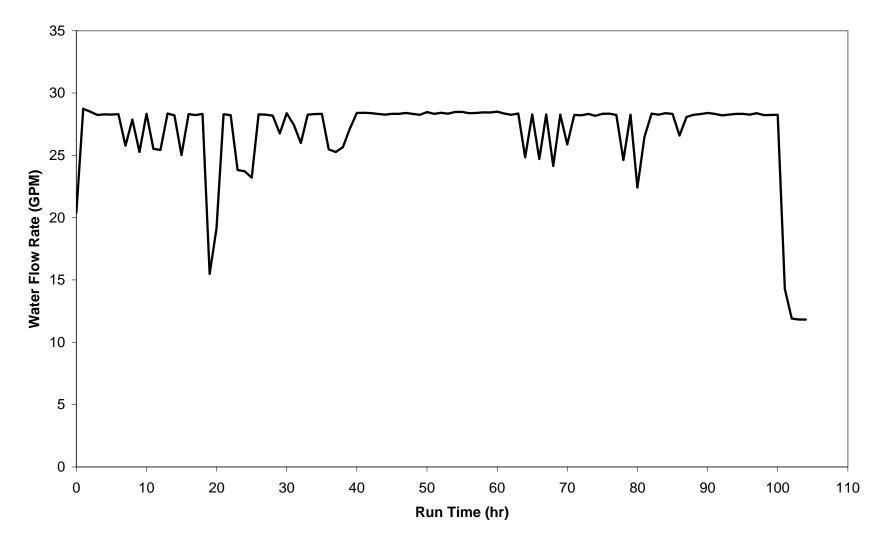


Figure 5.17.b. SBS cooling coil/jacket water flow rate (hourly average values) during Tests 2 and 3.

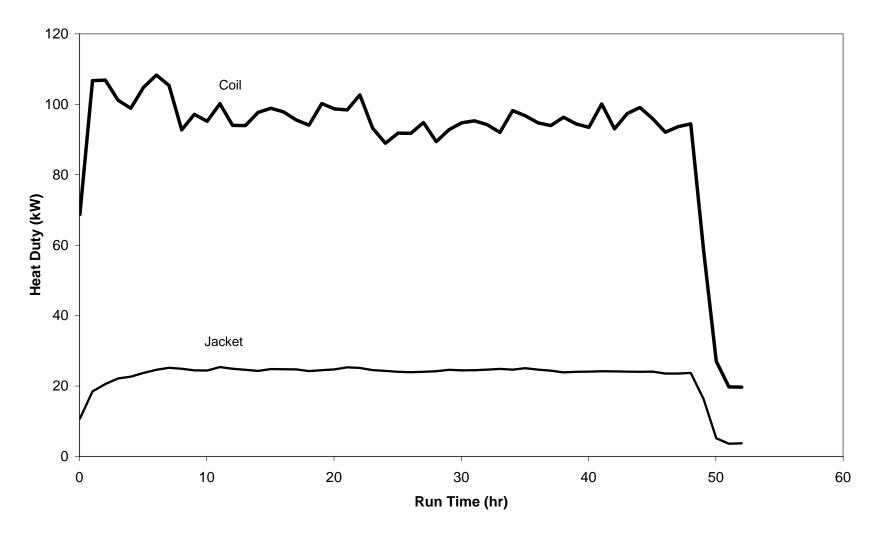


Figure 5.18.a. Calculated heat loads on the inner coil and jacket (hourly average values) during Test 1.

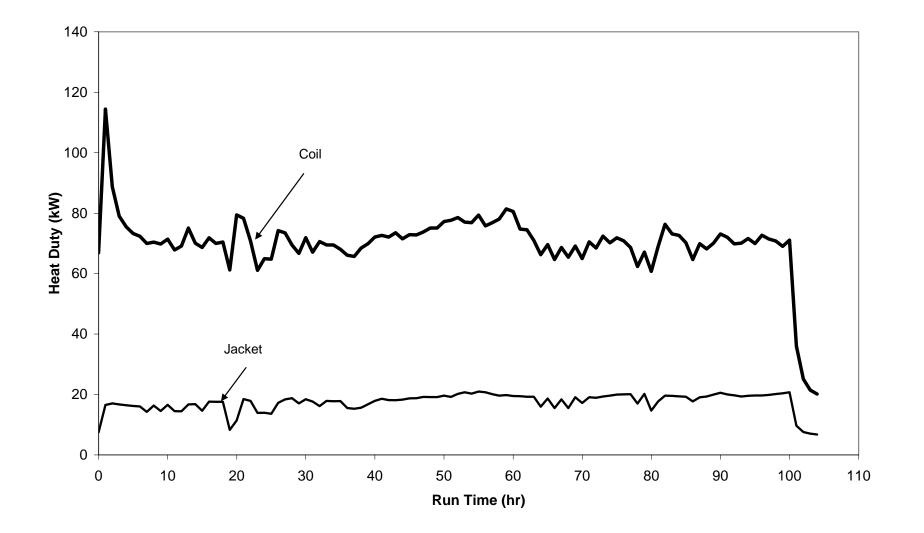


Figure 5.18.b. Calculated heat loads on the inner coil and jacket (hourly average values) during Tests 2 and 3.

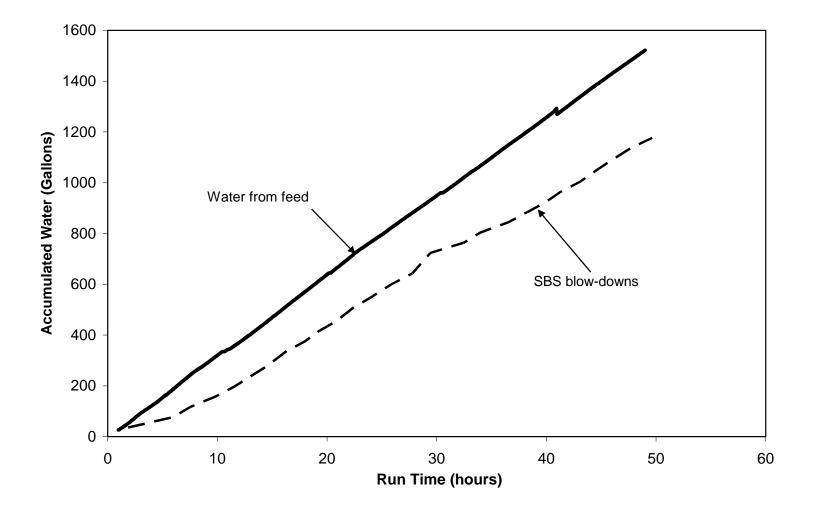


Figure 5.19.a. Accumulated SBS blowdown volume and accumulated feed water during Test 1.

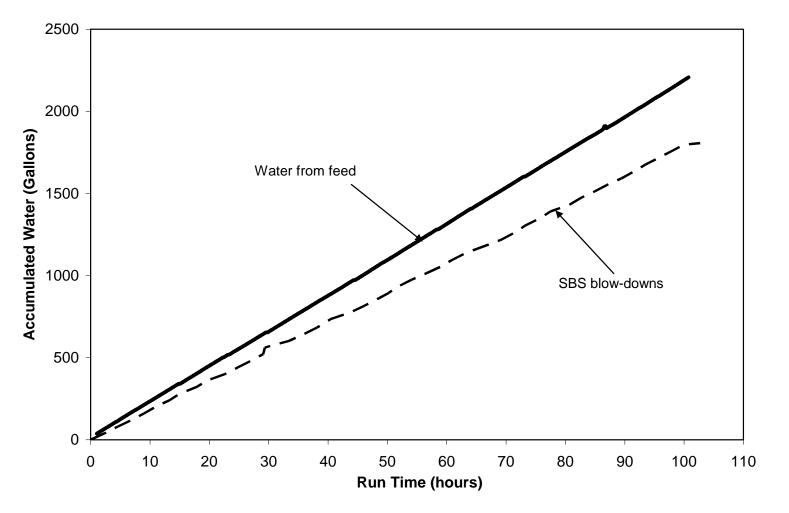


Figure 5.19.b. Accumulated SBS blowdown volume and accumulated feed water during Tests 2 and 3.

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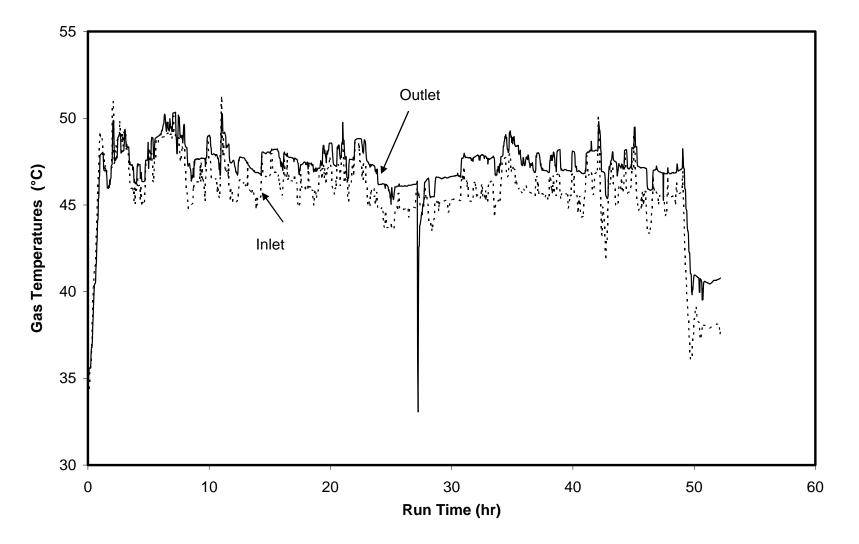


Figure 5.20.a. WESP inlet and outlet gas temperatures during Test 1. (Note: downward outlet temperature spikes are the result of WESP deluges.)

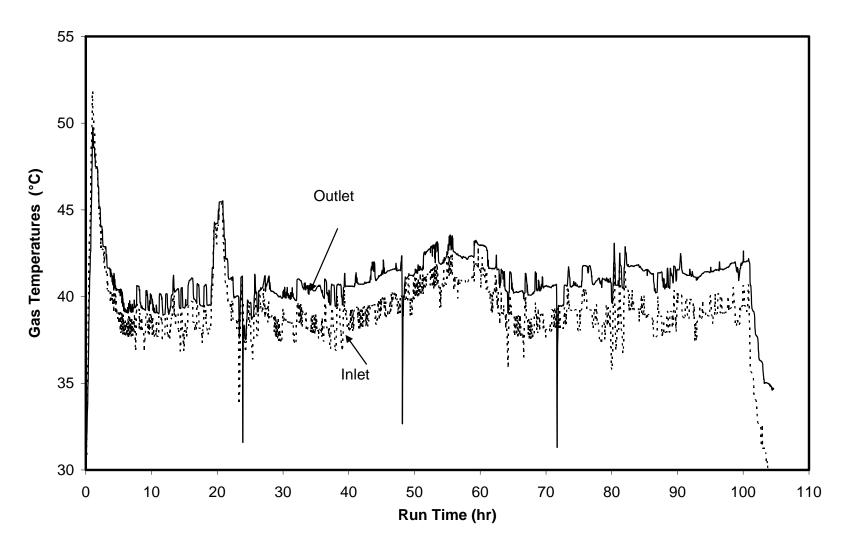


Figure 5.20.b. WESP inlet and outlet gas temperatures during Tests 2 and 3. (Note: downward outlet temperature spikes are the result of WESP deluges.)

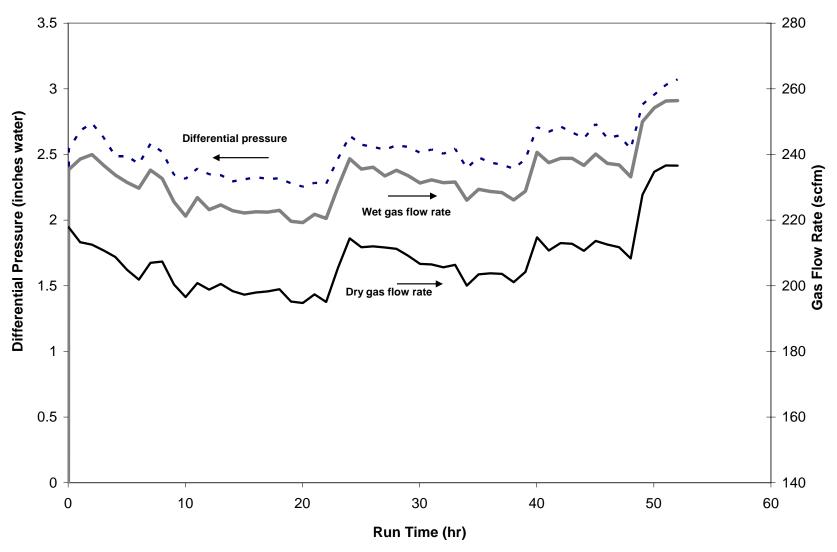


Figure 5.21.a. WESP differential pressure and outlet gas flow rate (hourly average values) during Test 1.

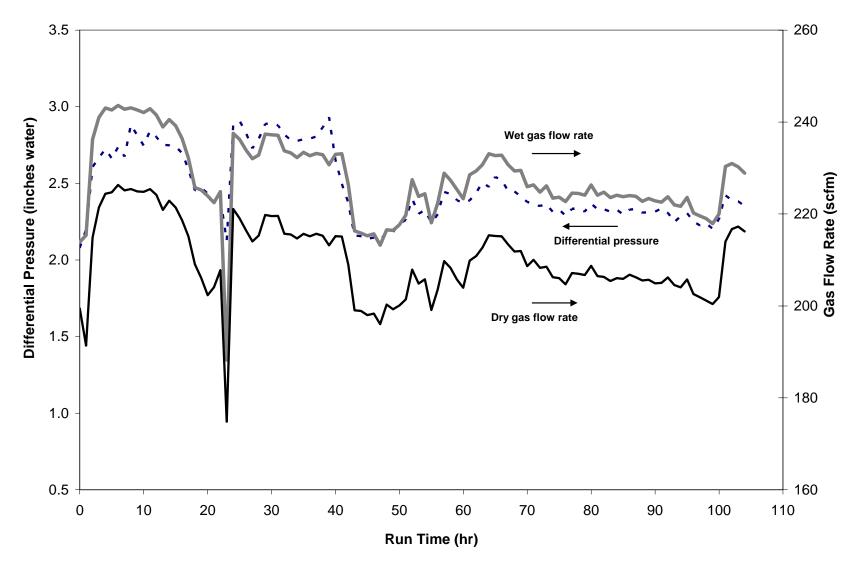


Figure 5.21.b. WESP differential pressure and outlet gas flow rate (hourly average values) during Tests 2 and 3.

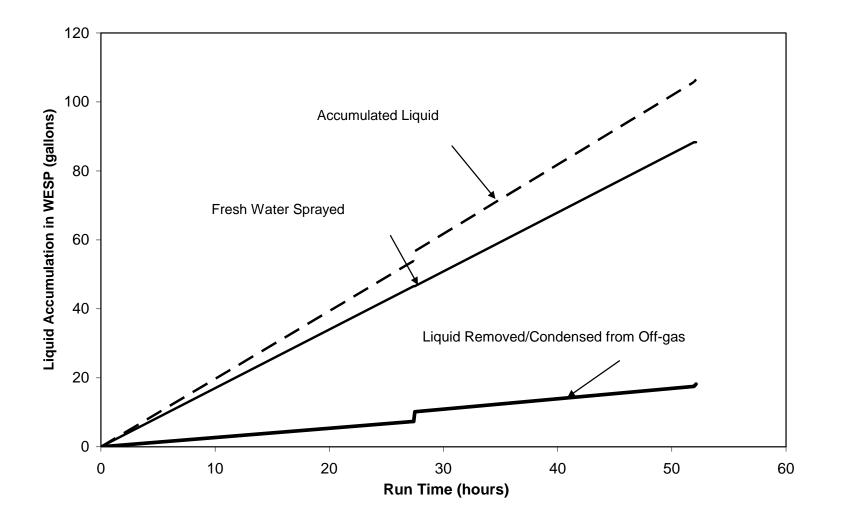


Figure 5.22.a. Accumulated WESP blowdown volume, accumulated fresh spray water, and water removed from off-gas during Test 1.

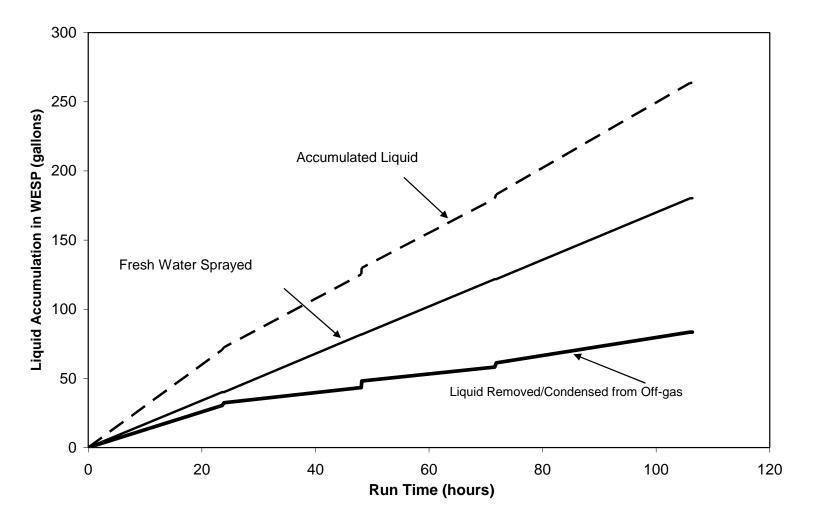


Figure 5.22.b. Accumulated WESP blowdown volume, accumulated fresh spray water, and water removed from off-gas during Tests 2 and 3.

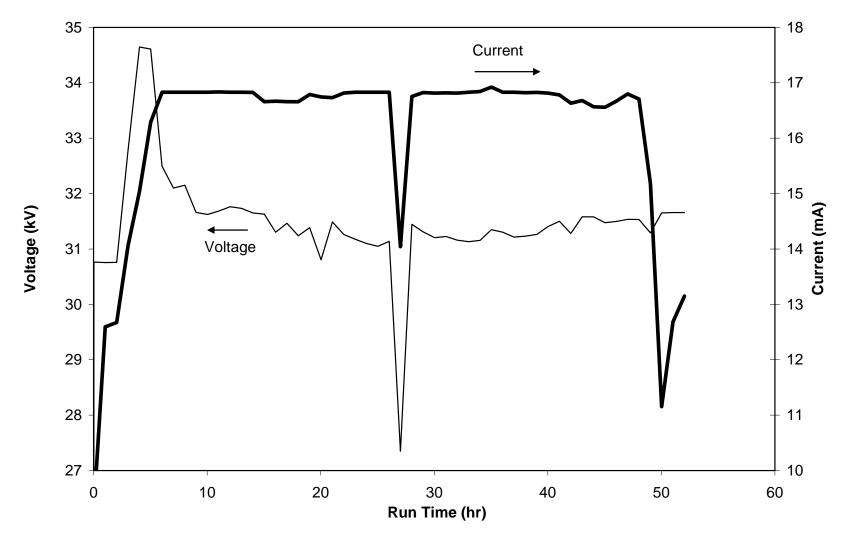


Figure 5.23.a. Voltage and current across the WESP during Test 1.

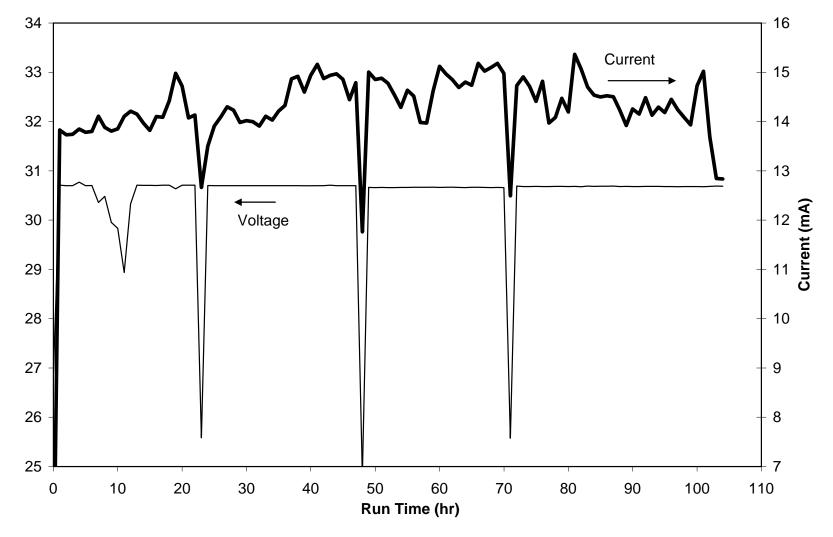


Figure 5.23.b. Voltage and current across the WESP during Tests 2 and 3.

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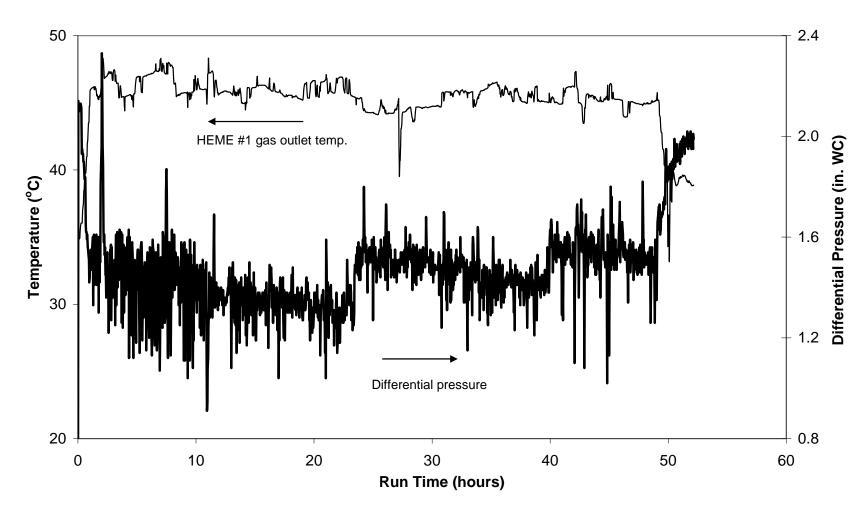


Figure 5.24.a. Outlet gas temperature and differential pressure for HEME #1 during Test 1.

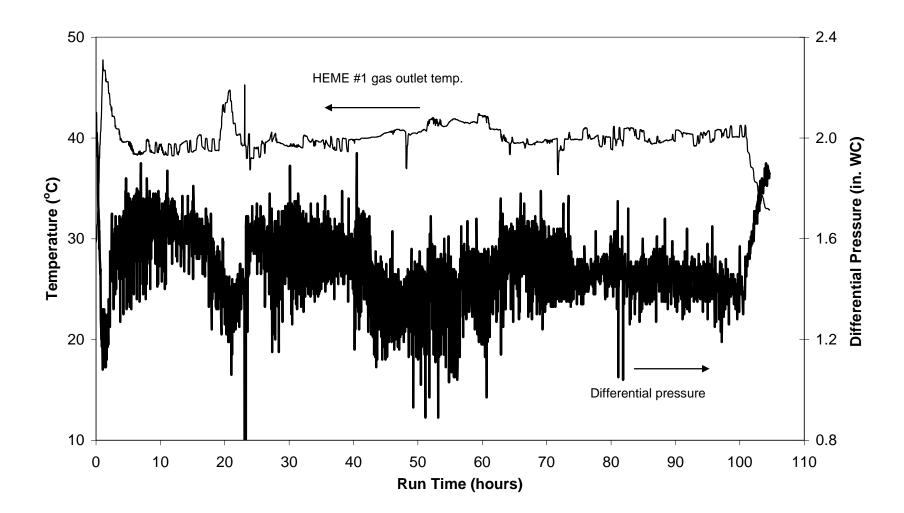


Figure 5.24.b. Outlet gas temperature and differential pressure for HEME #1 during Tests 2 and 3.

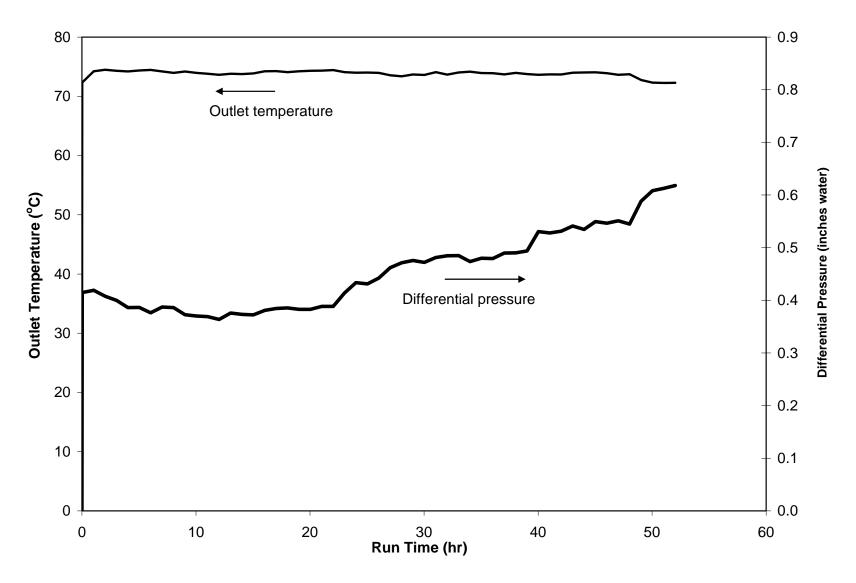


Figure 5.25.a. Outlet temperature and differential pressure for HEPA #1 (hourly average values) during Test 1.

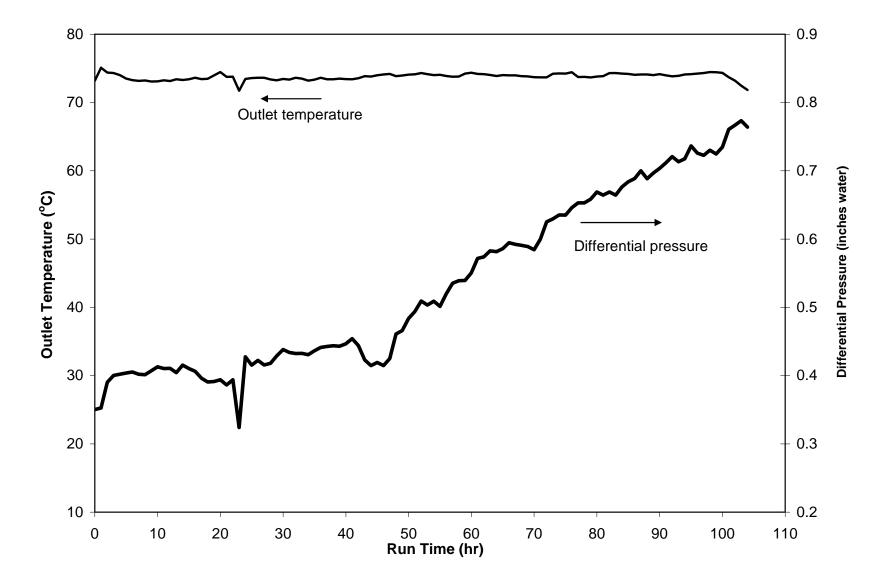


Figure 5.25.b. Outlet temperature and differential pressure for HEPA #1 (hourly average values) during Tests 2 and 3.

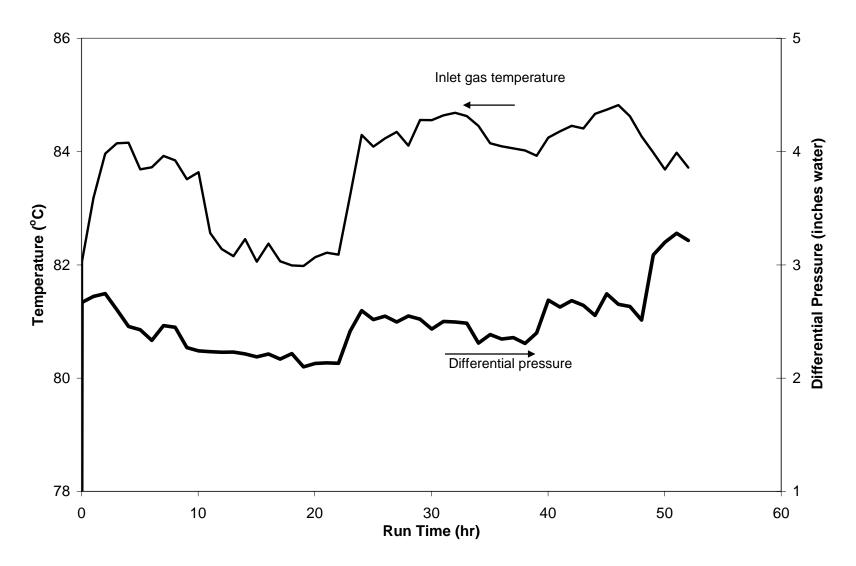


Figure 5.26.a. Inlet gas temperature and differential pressure for PBS (hourly average values) during Test 1.

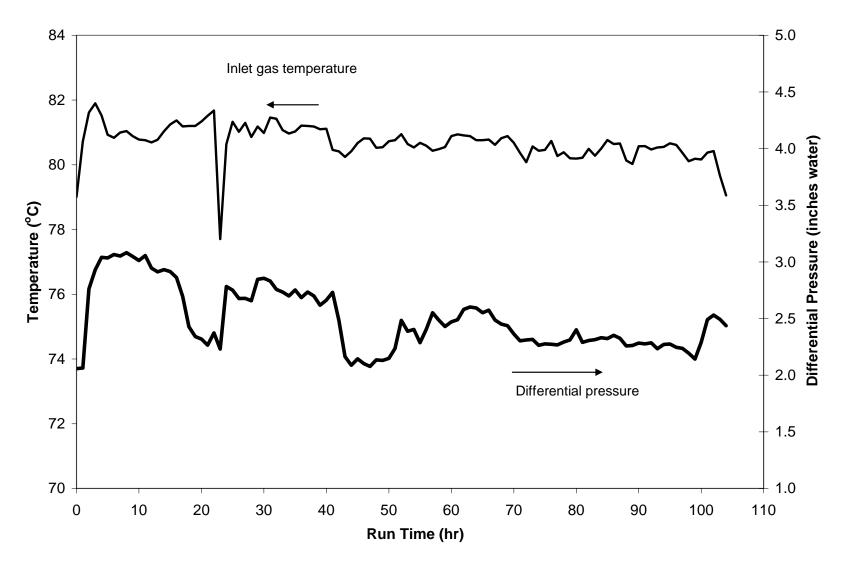


Figure 5.26.b. Inlet gas temperature and differential pressure for PBS (hourly average values) during Tests 2 and 3.

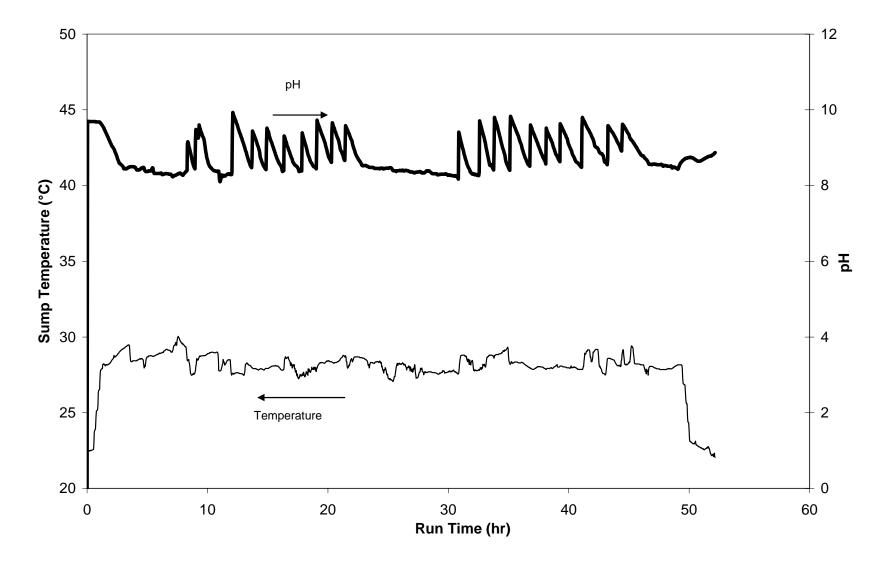


Figure 5.27.a. Sump temperature and pH for PBS (hourly average values) during Test 1.

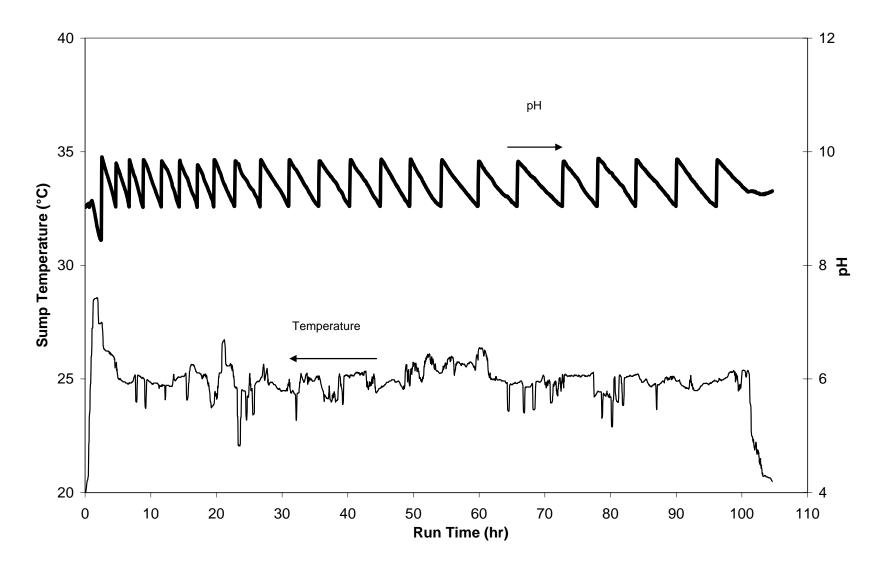
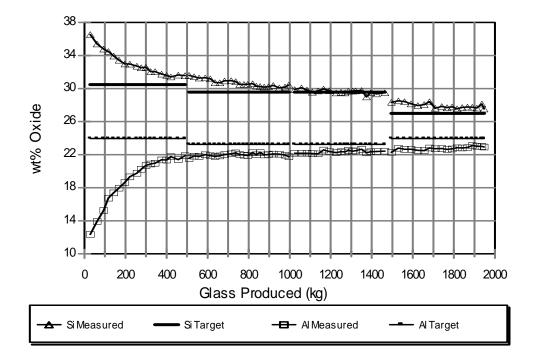
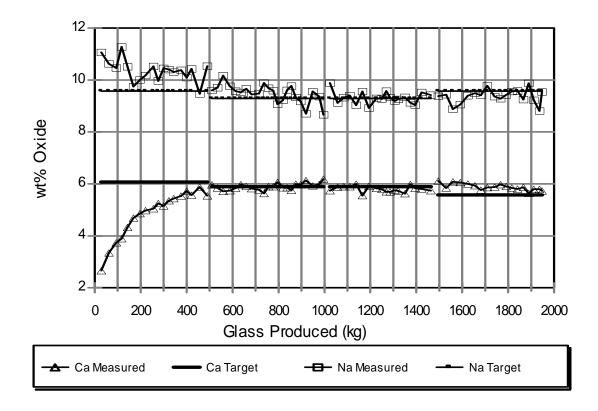
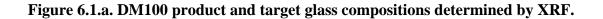


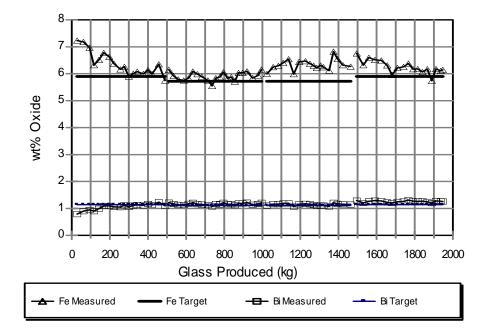
Figure 5.27.b. Sump Temperature and pH for PBS (hourly average values) during Tests 2 and 3.

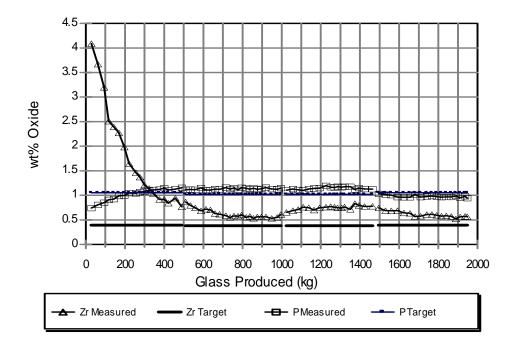
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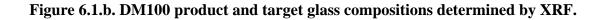












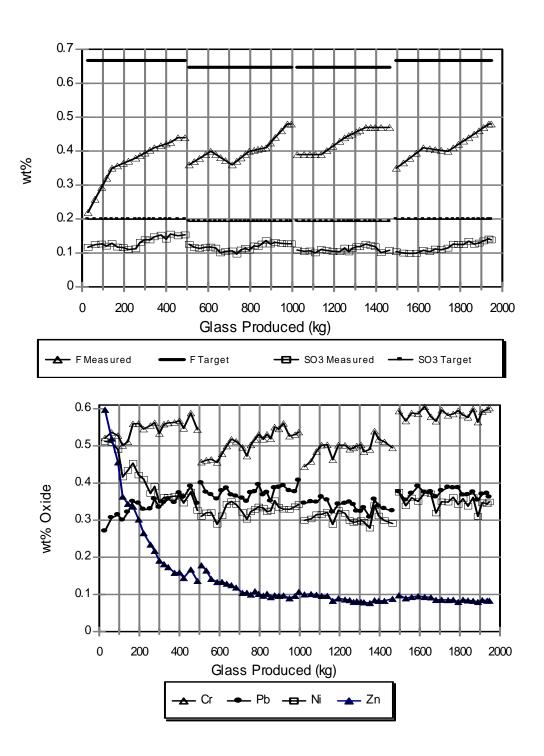
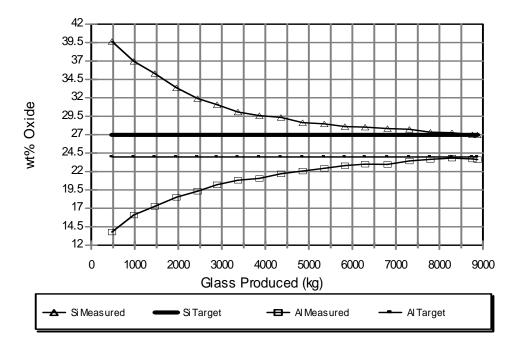


Figure 6.1.c. DM100 product and target glass compositions determined by XRF.



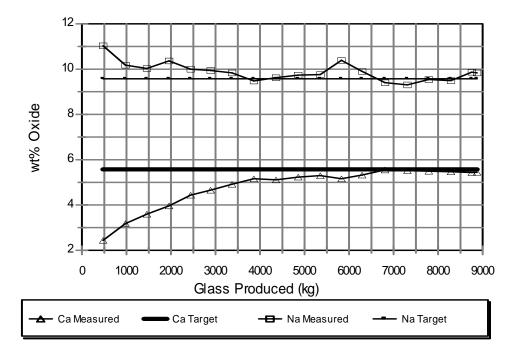
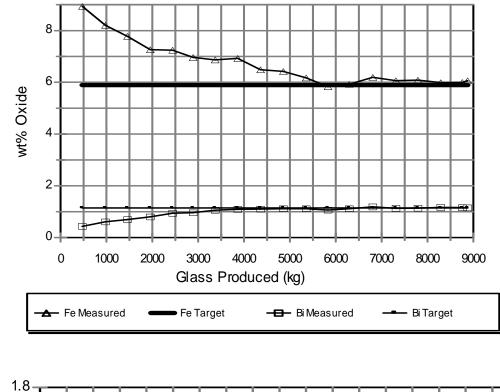
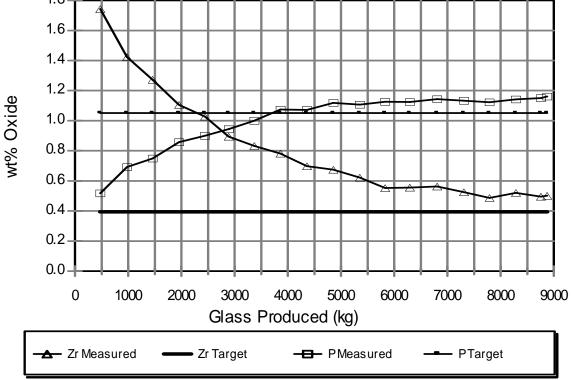
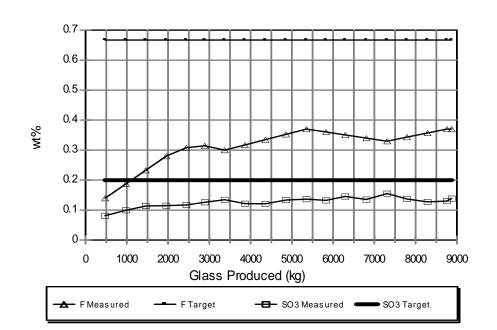


Figure 6.2.a. DM1200 product and target glass compositions determined by XRF.









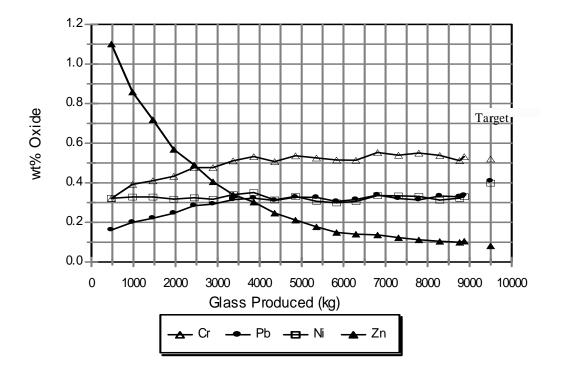
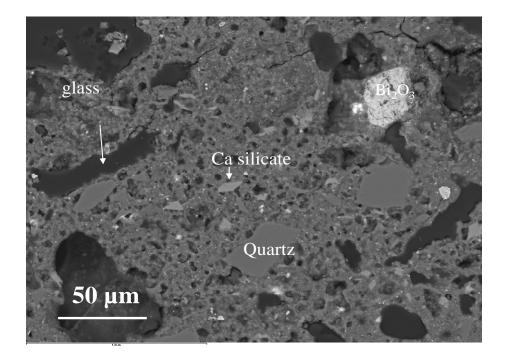
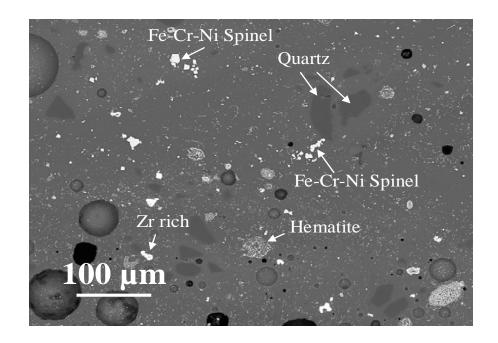


Figure 6.2.c. DM1200 product and target glass compositions determined by XRF.



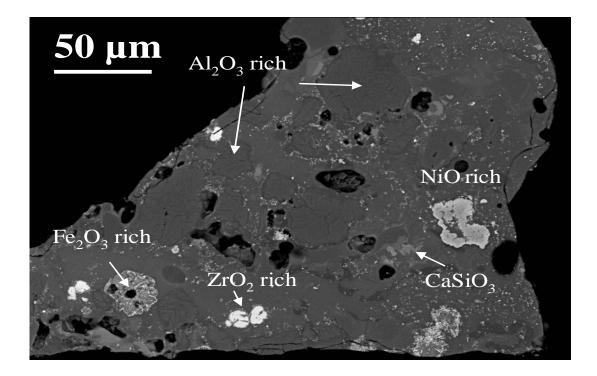
Backscattered electron image of a piece of cold cap sample (BLZ-O-37A)

Figure 6.3.a. SEM image of cold cap sample from DM100 Test 8.



Backscattered electron image of cold cap sample (BLZ-O-35A)

Figure 6.3.b. SEM image of cold cap sample from DM100 Test 8.



Backscattered electron image of the cold cap sample (O-12J-106A)

Figure 6.4. SEM image of cold cap sample from DM1200 Test 1.

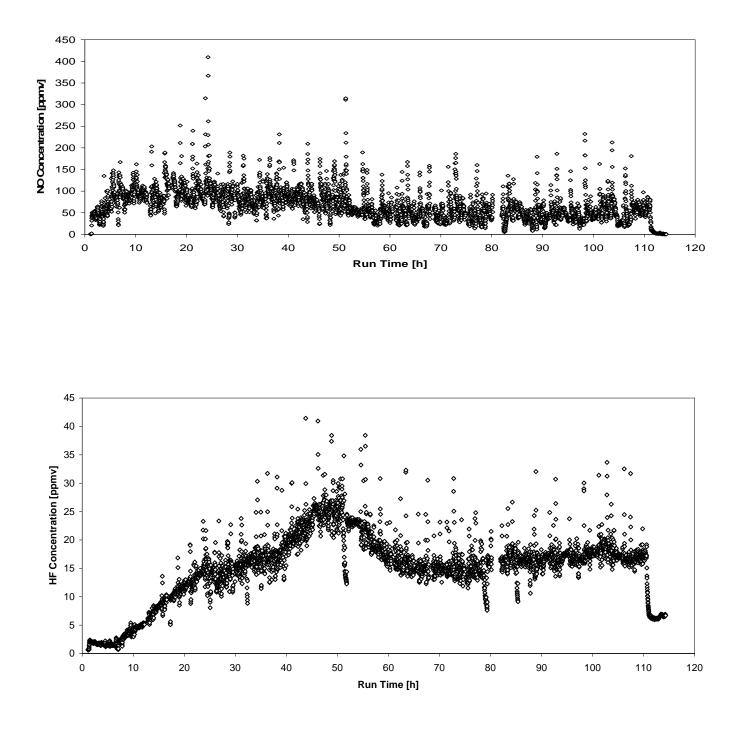


Figure 7.1. FTIR monitored NO and HF emissions during DM100 Tests 1 and 2.

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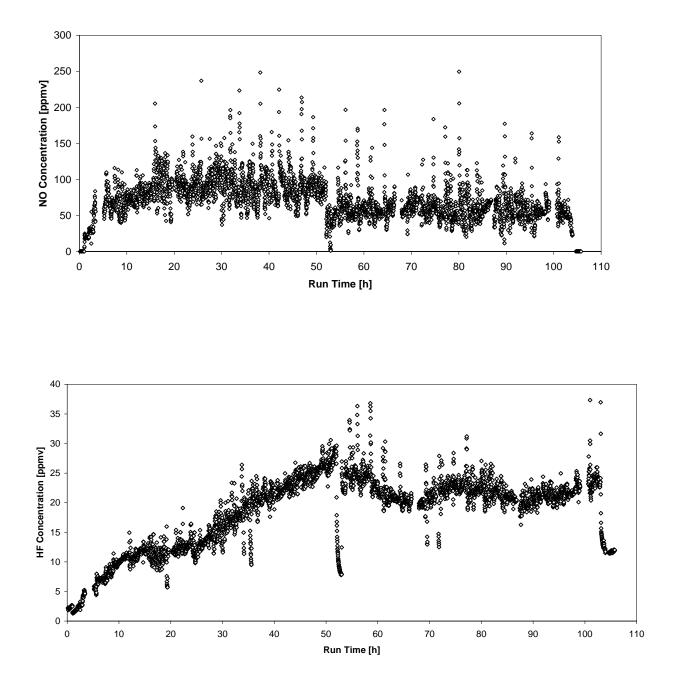


Figure 7.2. FTIR monitored NO and HF emissions during DM100 Tests 3 and 4.

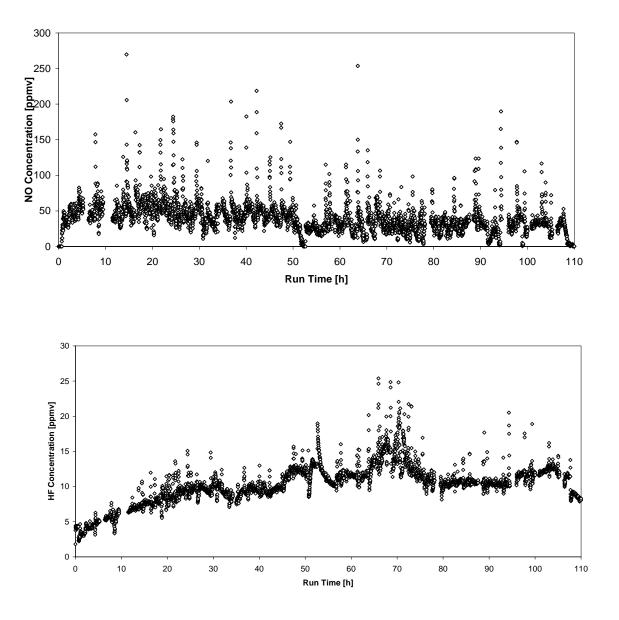


Figure 7.3. FTIR monitored NO and HF emissions during DM100 Tests 5 and 6.

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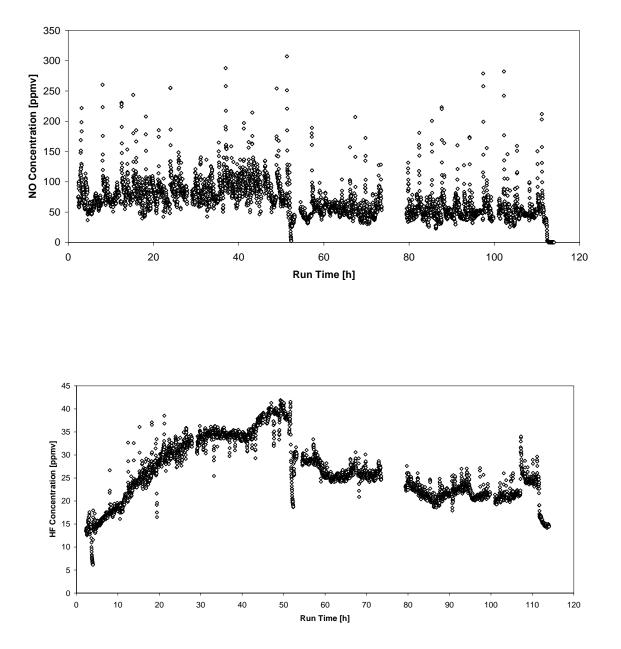


Figure 7.4. FTIR monitored NO and HF emissions during DM100 Tests 7 and 8.

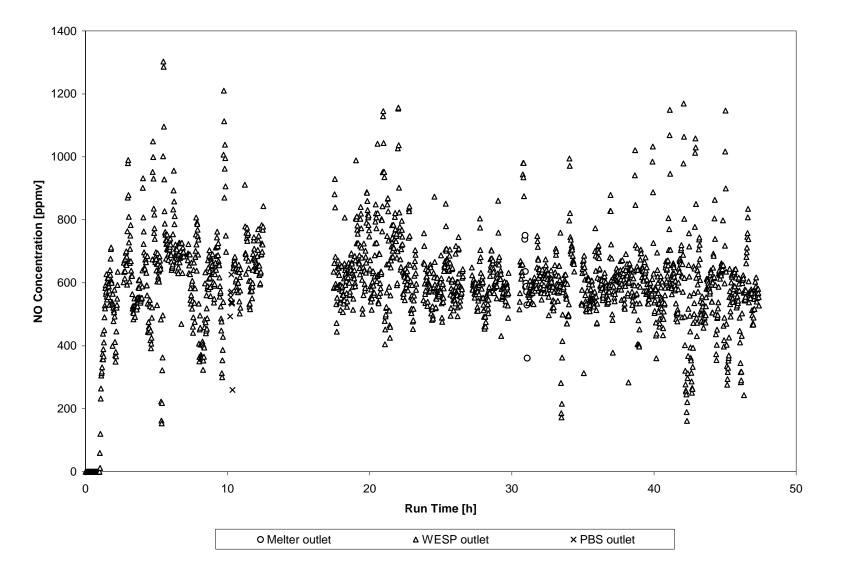


Figure 7.5. FTIR Monitored NO emissions during DM1200 Test 1.

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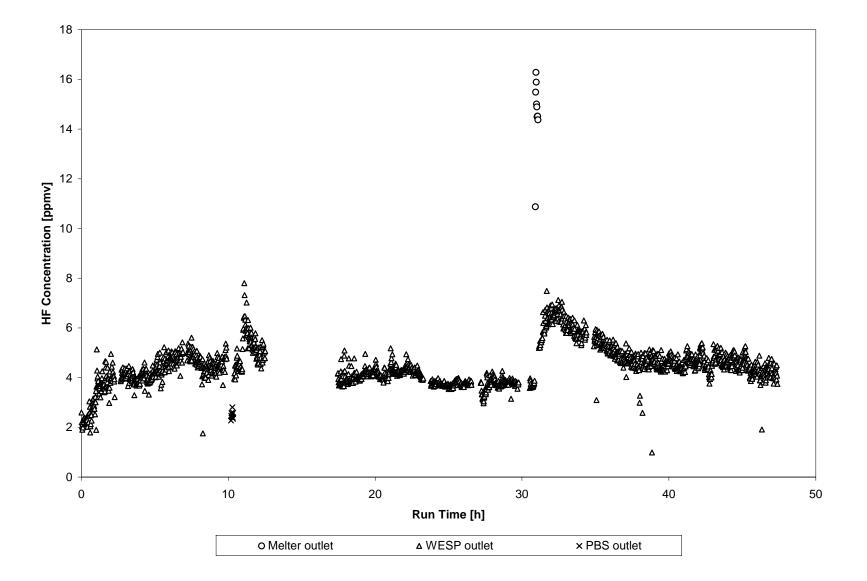


Figure 7.6. FTIR monitored HF emissions during DM1200 Test 1.

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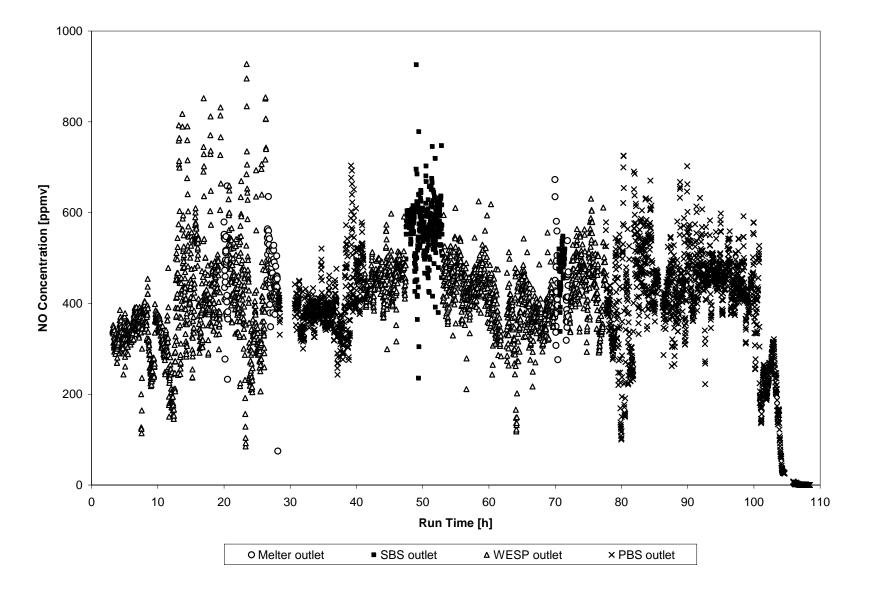


Figure 7.7. FTIR monitored NO emissions during DM1200 Tests 2 and 3.

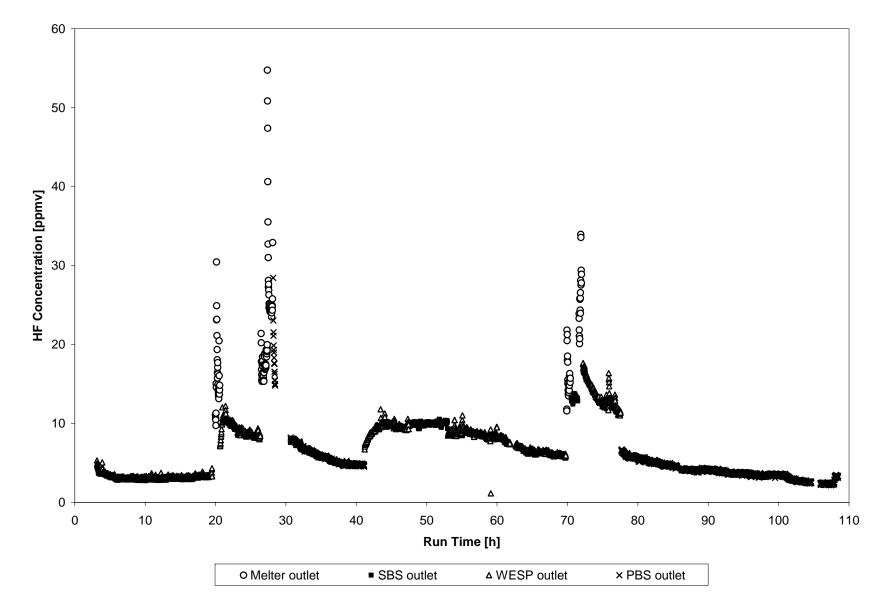


Figure 7.8. FTIR monitored HF emissions during DM1200 Tests 2 and 3.