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Tank Waste Remediation System Phase I High-Level Waste Feed Processability Assessment Report

S. L. Lambert, G. E. Stegen, and J. D. Vienna (PNNL)
Westinghouse Hanford Company, Richland, WA 99352
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Abstract: This report evaluates the effects of feed composition on the Phase I high-level waste immobilization process and interim storage facility requirements for the high-level waste glass. Several different Phase I staging (retrieval, blending, and pretreatment) scenarios were used to generate example feed compositions for glass formulations, testing, and glass sensitivity analysis. Glass models and data from laboratory glass studies were used to estimate achievable waste loading and corresponding glass volumes for various Phase I feeds. Key issues related to feed processability, feed composition, uncertainty, and immobilization process technology are identified for future consideration in other tank waste disposal program activities.

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**TANK WASTE REMEDIATION
SYSTEM PHASE I HIGH-LEVEL WASTE
FEED PROCESSABILITY
ASSESSMENT REPORT**

August 1996

S. L. Lambert
G. E. Stegen
Westinghouse Hanford Company
Richland, Washington

J. D. Vienna
Pacific Northwest National Laboratory
Richland, Washington

Prepared for
U.S. Department of Energy
Richland, Washington

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EXECUTIVE SUMMARY**ESI.0 INTRODUCTION**

This document examines the high-level waste (HLW) volume and composition sensitivities to a range of process retrieval and glass formulation options. The planned process for HLW delivery to the Phase I plants is to provide separate water washed feeds from tanks 241-AZ-101, 241-AZ-102, and 241-AY-102 (Bacon 1996). Tank 241-AY-102 will contain a mixture of tank 241-C-106 and 241-AY-102 wastes. For glass formulation purposes, various blends were also considered, including a nominal blend of these wastes consisting of 80 percent of Tank 241-AZ-101, 80 percent of Tank 241-AZ-102, 37 percent of Tank 241-C-106, and 15 percent Tank 241-AY-102 sludges. Feed composition estimates were primarily derived from the analysis of core samples and water washing tests performed on these samples. Glass models and data from laboratory glass studies were used to estimate the achievable waste oxide loading and corresponding glass volume for various Phase I feeds. This study also takes into consideration possible variations in feed composition due to uncertainties in the waste composition, blending, or pretreatment processes that will be applied to these wastes.

Glasses have been formulated for a range of anticipated Phase I wastes using estimates of their property data based on property models. This includes glass property data correlations developed by Pacific Northwest National Laboratory that have been incorporated into computer codes (models) to facilitate calculation of estimated glass properties. These models were extensively used in the current evaluation.

Key issues related to feed processability, feed composition uncertainty, and immobilization process technology are identified for future consideration in other tank waste disposal program activities.

ES2.0 GLASS DATA

Based on current models, iron and aluminum are the main components that control the volume and composition of the Phase I HLW glasses. Liquidus temperature and precipitation of nepheline ($\text{NaAlSi}_2\text{O}_6$) in the canister glass are the limiting properties of interest for most of these glasses. Liquidus temperature is important because this is the temperature where crystals may precipitate from the glass and clog the pour spout or possibly short the melter electrodes (i.e., if the sludge phases are electrically conductive). Precipitation of nepheline during cooling of the glass filled canisters is important because this is likely to produce an unacceptable waste glass due to a substantial decrease in durability and a diminished amount of aluminum in the remaining glass phase. Some data were found to be lacking with respect to the glass composition region of interest for the Phase I HLW feeds. Pacific Northwest National Laboratory provided glass formulation support by experimentally evaluating properties of glasses in this composition region. Unpublished glass property data were also obtained from Westinghouse Savannah River Company. These data were used to develop improved data correlations and calculational models for the Phase I wastes. Glass work produced significant results in the following areas:

- Significant component interaction effects were noted in several Phase I glasses. The most important interaction is the effect of aluminum on the solubility of iron (i.e., reducing aluminum increases the solubility of iron). Aluminum and iron have a dominant effect on the liquidus temperature of the Phase I HLW glasses, while aluminum, silica, and sodium control the nepheline precipitation behavior of these glasses.*
- New liquidus temperature models were developed from the current database and they are to be validated with a new set of glasses with targeted variations in Al_2O_3 , B_2O_3 , Cr_2O_3 , Fe_2O_3 , Li_2O , MgO , MnO , Na_2O , and SiO_2 to measure the effect of each component on the liquidus temperature of spinel. Spinel structures*

are represented by the formula $M^{\text{II}}M^{\text{III}}_2\text{O}_4$, where M^{II} consists of di-valent Fe, Mg, or Mn and M^{III} tri-valent Fe, Cr, or Al.

- *Precipitation of nepheline appears to be a significant issue for certain feeds. Significant progress was made in fiscal year 1996 in determining the composition ranges where nepheline precipitation could be a problem and estimating the possible durability effects of nepheline precipitation.*
- *Glass property models may be used to formulate and estimate the volume of HLW glasses. Based on these models and glass property data, the maximum waste loading for the Phase I glasses varies from 29.3 to 30.7 wt% on an adjusted feed oxide basis. The adjusted feed oxide basis is defined as the weight of total non-volatile feed oxides excluding sodium and silica on an oxide basis divided by weight of the product glass. This indicates there is a 17 percent contingency or operating margin between the maximum theoretical waste loading and the minimum 25 percent adjusted feed oxide basis specified in the Request for Proposal. The 25 percent adjusted feed oxide appears to be a conservative and achievable value for Phase I.*

The primary conclusions from these findings are as follows:

- *There is an incentive to reduce the amount of aluminum in the Phase I waste with caustic washing.*
- *The iron solubility limit in the HLW glass may be reduced significantly by high aluminum concentration due to liquidus temperature constraints. High concentrations of chromium and manganese also have an adverse effect on allowable iron compositions because of liquidus temperature effects.*

- *The nepheline region can be avoided by defining a minimum limit for SiO_2 such that glasses with a normalized SiO_2 content ≥ 0.62 (based on the sum of $\text{SiO}_2 + \text{Na}_2\text{O} + \text{Al}_2\text{O}_3$) should not be susceptible to nepheline precipitation in the canister. The defined limit for nepheline is important for certain wastes, such as Tank 241-C-106 waste, where the waste oxide loading limits are likely to be determined by the nepheline forming characteristics of the glass.*
- *Glass property data and related models should be considered in developing future baseline glass quantity estimates. Recent glass data identify some important component interaction effects that are not well represented in simplified single component and multi-component constraint limits. A primary example is the effect of aluminum on iron solubility in glass. This interaction generally results in a larger volume of glass and more incentive for caustic washing as compared to glass estimates derived from single component or multi-component limits.*

ES3.0 GLASS VOLUME AND WASTE OXIDE LOADING ESTIMATES

In this study, several different Phase I staging (retrieval, blending, and pretreatment) scenarios were used to generate example feed compositions for glass formulations, testing, and glass sensitivity analysis. It should be recognized that these scenarios will not necessarily match the final process that is selected for Phase I. Several glasses were formulated for each waste based on glass property models to establish the range of possible waste oxide loading for the wastes identified in the Phase I Request for Proposal.

The glass models and glass property results were used to produce the estimates in Table ES-1 for the water washed (base case) and caustic washed Phase I wastes (Tanks 241-AZ-101, 241-AZ-102, and 241-C-106 wastes and various blends of these wastes). The glass models include the Fulcher first-order viscosity model (Section 5.3), spinel second-order liquidus temperature model (Section 5.5), second-order product consistency test B and Na release models and the nepheline model (Section 5.6).

Table ES-1. Glass Volume Estimates for Unblended Phase I Wastes.

Phase I tanks	Water washed waste	Waste inventory ^a (MT)	Base glass WOL, wt %	Base glass AFO, wt %	Cans of base ^b glass (limit)	Cans w/50% Al ₂ O ₃ removed (limit)	Cans w/75% Al ₂ O ₃ removed (limit)
241-AZ-101	241-AZ-101	109.46	36.08	30.26	183 (T _L)	154 (T _L)	145 (T _L)
241-AZ-102	241-AZ-102	172.99	34.40	29.28	305 (T _L)	265 (T _L)	268 (T _L)
241-C-106	241-C-106A	485.52	67.45	30.65	436 (neph.)	356 (T _L /neph)	316 (low vis)
Total		767.97			924	775	729

AFO = Adjusted feed oxide

WOL = Waste oxide loading

^aWater washed waste oxide inventory assuming 100% retrieval efficiency.

^bDefense Waste Processing Facility canisters of glass limited by T_L liquidus temperature or nepheline.

If caustic washing processes are employed, the glass models predict that 80 canisters could be saved by removing 50 percent of the Al₂O₃ from Tank 241-C-106 waste and 120 canisters if 75 percent of the Al₂O₃ is removed from this waste. Caustic washing processes appear to be less desirable for Tanks 241-AZ-101 and 241-AZ-102 wastes, producing 29 and 40 fewer canisters, respectively. These estimates are based on the physical and chemical properties of the glass. Caustic washing studies will be necessary to determine whether it is possible to remove this much aluminum from the Phase I wastes. Also, it is unclear whether the Tank Waste Remediation System safety basis could be modified in time to allow enough tank space for a caustic washing option for these wastes.

Table ES-2. Phase I Blend Compositions.

<i>Nominal Blend:</i>	$0.8 * 241\text{-AZ-101} + 0.8 * 241\text{-AZ-102} + 0.37 * 241\text{-C-106} + 0.15 * 241\text{-AY-102}$
<i>AZ Blend:</i>	$241\text{-AZ-101} + 241\text{-AZ-102}$
<i>AYC Blend:</i>	$241\text{-C-106} + 241\text{-AY-102}$
<i>AYC-A Blend:</i>	$241\text{-AY-102} + 0.1625 * 241\text{-C-106}$
<i>AYC-B Blend:</i>	$241\text{-AY-102} + 0.4875 * 241\text{-C-106}$
<i>AZC-A Blend:</i>	$AZ\ Blend + 0.4875 * 241\text{-C-106}$
<i>AZC-B Blend:</i>	$AZ\ Blend + 0.1625 * 241\text{-C-106}$

Glass property data indicate the maximum waste oxide loading for the nominal blend feed (Table ES-2) is 48.5 wt% total oxides or 30.9 wt% on an adjusted feed oxide basis. If the Phase I wastes are processed separately, the average waste oxide loading is 30.1 wt% adjusted feed oxide (mass weighted average basis, Table ES-1). This indicates that only a small benefit will be gained from blending in terms of reducing the overall volume of glass. There may be other benefits from blending, including reducing the concentration of problem components such as silver and reducing the number of glasses and process conditions that need to be qualified as part of product acceptance.

The current estimates of waste glass volume do not include any allowance for operating margin contingency (i.e., to compensate for process fluctuations and uncertainties in the waste or glass composition analysis and glass property models). If it is assumed that production glasses are formulated to 85 percent of the maximum waste loading, the estimate waste loadings for the tank-by-tank and nominal blend feed are 25.6 and 26.3 percent adjusted feed oxide, respectively. For the tank-by-tank feed, this results in 957,000 kg of glass (i.e., 580 defense waste processing facility canisters) for the minimum Phase I feed of 245,000 kg adjusted feed oxide or 1,816,000 kg of glass (i.e., 1,100 defense waste processing facility canisters) for the maximum feed of 465,000 kg adjusted feed oxide.

ES4.0 NOBLE METALS

Noble metal oxides and alloys of silver, palladium, tellurium, rhodium, and RuO_2 are virtually insoluble in borosilicate glass. The Phase I water washed nominal blend contains 0.38 wt% noble metals (Ag_2O , RuO_2 , TeO_2 , PdO , and Rh_2O_3). At 48 percent waste oxide loading in the glass, the noble metals concentration would be 0.182 wt%. If the Phase I feeds are processed separately, Tanks 241-AZ-101, 241-AZ-102, and 241-C-106 glasses should nominally contain 0.17, 0.11, and 0.24 percent noble metals, respectively. Recent grab samples from Tank 241-C-106 indicate the bounding noble metals concentration could be as high as 0.52 wt% in glass because of the amount of silver and ruthenium in this waste. This noble metals concentration is likely to cause premature failure of the melter electrodes and bottom refractory (based on the low melting temperature of silver [960 °C] and laboratory-scale tests in the Research-Scale Melter performed with feeds containing 0.33 to 0.46 wt% noble metals). An added concern is the high silver inventory in Tank 241-AY-102, with 1.85 wt% silver oxide (Ag_2O) in the waste. If the noble metals concentration is determined to be too high, other wastes should be considered as potential blend stock for mitigating the high silver and ruthenium inventories in Tanks 241-C-106 and 241-AY-102. This issue mostly affects the performance of joule-heated melters, although noble metal deposits have also plugged the bottom drains of various induction-heated melters.

ES5.0 SENSITIVITY ANALYSIS

Sensitivity analyses were performed to evaluate the effects of feed composition and more aggressive caustic washing processes for the Phase I wastes. The purposes of this analysis are to provide information to the pretreatment and waste retrieval functions and to identify high priority issues to be addressed in any future waste characterization activities.

The most important HLW components from a glass making perspective are Al_2O_3 and Fe_2O_3 . Because glass volume predictions are important to overall project cost, a sensitivity

study was performed to assess the likely impact of varying the amount of Al_2O_3 and Fe_2O_3 in the Tanks 241-AZ-101 and 241-C-106 wastes. For Tank 241-AZ-101 waste, the Al_2O_3 to Fe_2O_3 ratio of 0.74 is an important characteristic of the glass. If this ratio is less than 0.74, the glass volume is controlled by the liquidus temperature characteristics of the glass, while the nepheline forming properties control the volume of glass above this value.

If the mass of all components except Al_2O_3 or Fe_2O_3 is held constant, the volume of glass from processing Tank 241-AZ-101 waste is projected to increase 4.3 and 5.6 percent, respectively, per 10 percent increase in the amount of Fe_2O_3 or Al_2O_3 . The total glass volume would decrease 0.65 and 2.8 percent, respectively, with a decrease of 10 percent in the amount of Fe_2O_3 or Al_2O_3 in this waste. Similar trends were also identified for Tank 241-C-106 waste, with an increase of 4.3 and 4.7 percent in glass volume per 10 percent increase in Fe_2O_3 or Al_2O_3 , respectively, and a decrease of 1.1 and 3.6 percent with a 10 percent decrease in the amount of Fe_2O_3 or Al_2O_3 in this tank. These figures can be used to determine the impact of over-estimating or under-estimating the mass of these components (based on a repository disposal fee of \$356,000 per canister). These results reflect the effect of reducing the quantity of Al_2O_3 in the waste by caustic washing or inadvertently adding Al_2O_3 by precipitation from the liquid phase.

ES6.0 PROCESS TECHNOLOGY

Additional information has recently become available on performance of the high-temperature melter and high-temperature glasses. Tests with a refractory-lined high-temperature melter indicate that the melt-rate capacity (at 1,350 °C) is only about 10 percent higher than the capacity of the low-temperature melter (at 1,150 °C). Dissolution of melter refractory components into glass increases aluminum and chromium and thus reduces the previously expected waste oxide loading advantage of the high-temperature melter (if corrosion product components are included in calculating the liquidus temperature limits for glass formulation). Glass formulation studies have also shown that the solubility of

chromium oxide (as one of several spinel-forming components) is only slightly affected by the temperature and composition of borosilicate and aluminum silicate glasses (thus reducing the potential advantage of the high-temperature melter). Rapid corrosion of the melter electrode materials at high-temperature ($T > 1,450\text{ }^{\circ}\text{C}$) also indicates that development of a high-temperature melter is likely to be more expensive than originally anticipated due to the need to identify and test more exotic (and expensive) electrode materials. For the Phase I program, the private vendors may elect to use other melters, such as the high-temperature cold crucible melter, that may not be susceptible to refractory corrosion and potential sludge accumulation problems discussed in this report.

Glass property data indicate that precipitation of sludge forming solids (spinel) limits the achievable waste oxide loading for sludges from Tanks 241-AZ-101/241-AZ-102. If a practical sludge removal system was developed for the low-temperature melter, it might be feasible to process glasses that contain a small fraction of undissolved solids (i.e., precipitated crystals). Glass formulation results indicate the waste oxide loading of pretreated sludges from Tanks 241-AZ-101/241-AZ-102 could be increased by 30 to 40 percent if the solids precipitation limits (liquidus temperature constraints) were relaxed. With this feature, glasses could be formulated beyond the liquidus temperature limit for the low-temperature melter to possibly match the waste loading capabilities of the high-temperature melter, thus reducing the predicted life-cycle costs for the low-temperature melter. If the liquidus temperature limit is raised from 1,050 to 1,100 $^{\circ}\text{C}$, approximately 53 fewer canisters (10.7 percent less glass) should be produced from the nominal blend feed.

Glass formulation models indicate that only two components (Fe_2O_3 and Al_2O_3) need to be known with reasonable precision (to within 15 percent or their true values) to estimate the amount of glass likely to be produced from Tank 241-AZ-101 waste, and four components (SiO_2 , Na_2O , Fe_2O_3 , and Al_2O_3) to estimate the volume of glass from Tank 241-C-106 waste. Other components, from a glass formulation perspective, that need to be known (to within 100 percent of their true values) are Na_2O , MnO , and ZrO_2 for Tank 241-AZ-101 waste and Cr_2O_3 , MnO , and CaO for Tank 241-C-106 waste. While it may be desirable to obtain as

much information as possible, the volume and composition of most glasses can be reasonably determined from Fe_2O_3 and Al_2O_3 composition data, together with some knowledge of Cr_2O_3 , MnO , noble metals (Ag_2O , PdO , Rh_2O_3 , and Ru_2O_3), and semi-soluble waste components (P_2O_5 and SO_4).

This report also discusses the performance characteristics of the low- and high-temperature melters, and radioactive test melters and test facilities at the Hanford Site. Similar information is provided on the glass property constraints (i.e., certain waste form qualification requirements, viscosity, electrical conductivity, liquidus temperature, and durability) and on the glass solubility constraints for minor components (Cr, P, Cl, F, S, Bi, U, noble metals, and Ti). Balance of plant issues are also discussed, including the possible effects of Hg, Cd, Cl, F, SO_4 , NO_x , and criticality control.

CONTENTS

1.0 INTRODUCTION	1
1.1 BACKGROUND	1
1.2 PURPOSE	2
1.3 SCOPE	3
2.0 PHASE I HIGH-LEVEL WASTES	5
2.1 SELECTION OF PHASE I WASTES	5
2.2 SOURCE OF WASTES	5
2.3 GENERAL CHARACTERISTICS OF WASTES	6
3.0 CURRENT ESTIMATES OF MELTER FEED COMPOSITION	11
3.1 241-AZ-101 TANK WASTE	11
3.2 241-AZ-102 TANK WASTE	12
3.3 241-AY-102 TANK WASTE	12
3.4 241-C-106 TANK WASTE	12
3.5 NOMINAL BLEND COMPOSITION	20
4.0 VITRIFICATION	23
4.1 IMPACT OF VITRIFICATION TECHNOLOGY ON FEED PROCESSABILITY	23
4.2 STATUS OF VITRIFICATION TECHNOLOGY	23
4.3 HIGH-LEVEL WASTE MELTERS	24
4.3.1 Low-Temperature, Joule-Heated Melter	24
4.3.2 High-Temperature, Joule-heated Melter	33
4.3.3 Radioactive Melter Tests	44
5.0 FEED PROCESSABILITY	49
5.1 WASTE FORM/WASTE FORM QUALIFICATION	49
5.2 GLASS PROPERTY CONSTRAINTS/MELTER FEED COMPOSITION CONSTRAINTS	52
5.3 VISCOSITY	54
5.4 ELECTRICAL CONDUCTIVITY	56
5.5 LIQUIDUS TEMPERATURE	56
5.6 DURABILITY	67
5.7 GLASS SOLUBILITY CONSTRAINTS	76
5.7.1 Chromium Oxide Solubility	76
5.7.2 Phosphorus Solubility	79
5.7.3 Chloride and Fluoride Solubility	81
5.7.4 Sulfate Solubility	82
5.7.5 Bismuth and Uranium Solubility	83
5.7.6 Noble Metals Solubility	84
5.7.7 Titanium Solubility	87
5.7.8 Minor Component Solubility Limits	88

5.8	COMPARISON OF SIMULANTS TO RADIOACTIVE PHASE I WASTES	89
5.8.1	Slurry Properties	90
5.8.2	Glass Properties	91
6.0	PROCESSABILITY ASSESSMENT FOR PHASE I FEEDS	95
6.1	WASTE COMPOSITIONS FOR GLASS FORMULATION DEVELOPMENT	95
6.2	GLASS FORMULATION	98
6.2.1	Glass Formulation Development for the Water Washed Nominal Blend	104
6.2.2	Glass Formulation Development for the Caustic Washed Nominal Blend	107
6.2.3	Glass Formulation development for the Water Washed AZ Blend	108
6.2.4	Glass Formulation Development for the Water Washed 241-C-106A Tank Waste	109
6.2.5	Glass Formulation Development for the Water Washed 241-C-106B Tank Waste	110
6.3	GLASS VOLUME PREDICTIONS	111
6.3.1	Bounding Composition Limits for the (Water Washed) Nominal Blend Glass	111
6.3.2	Bounding Composition Limits for the (Caustic Washed) Nominal Blend Glass	113
6.3.3	Bounding Composition Limits for the AZ Blend Glass	115
6.3.4	Bounding Composition Limits for the 241-C-106A Glass	117
6.3.5	Bounding Composition Limits for the 241-C-106B Glass	119
6.3.6	Summary of Results for Phase I Experimental Glasses	120
6.3.7	Glass Volume Estimates for Phase I Wastes	123
6.3.8	Bounding Effects of Al_2O_3 and Fe_2O_3 in Glass	125
6.3.9	Glass-Limiting Components in the Waste	139
6.4	BALANCE OF PROCESS	143
6.4.1	Mercury	143
6.4.2	Cadmium	144
6.4.3	Chloride, Fluoride, and Sulfate	145
6.4.4	Nitrogen Oxides	147
6.4.5	Criticality Control	148
7.0	RESULTS AND CONCLUSIONS	151
7.1	GLASS COMPOSITION	152
7.2	MELTER PERFORMANCE SENSITIVITIES	152
7.3	MELTER TECHNOLOGY	153
7.4	PHASE I GLASS PROPERTIES	154
7.5	GLASS SOLUBILITY LIMITS	156
7.6	WASTE OXIDE LOADING AND GLASS VOLUME ESTIMATES	157
7.7	BALANCE OF PROCESS	165

8.0 REFERENCES	169
--------------------------	-----

LIST OF FIGURES

2-1. Composition of the Principal Streams from the Uranium Recovery Process.	8
2-2. Composition of the Principal Streams from PUREX and B Plant Processes.	9
5-1. Predicted Versus Measured Liquidus Temperature of Zr-Containing Crystals for the First-Order Mixture Model	60
5-2. Predicted Versus Measured Liquidus Temperature of Spinel Phases in Composition Variability Study Glasses Based on First-Order Mixture Model . . .	61
5-3. Comparison of Normalized Boron Release From Quenched and Canister Centerline Cooled Glass.	69
5-4. Na ₂ O - Al ₂ O ₃ - SiO ₂ Phase Diagram	71
5-5. Nepheline Region in Na ₂ O - Al ₂ O ₃ - SiO ₂ Phase Diagram	72
5-6. Effect of Neutralized Current Acid Waste Loading on 7-Day Product Consistency Test Normalized Element Sodium and Silicon Releases	74
5-7. Effect of Aluminum Oxide on Normalized Boron Release from Composition Variability Study and HW39-4 Glasses	75
6-1. Effect of Liquidus Temperature, Nepheline, and Composition Variability Study Limits on Volume of Glass versus Al ₂ O ₃ in Tank 241-AZ-101 Waste	128
6-2. Effect of High-Level Waste Glass Model and Composition Variability Study Limits on Volume of Glass versus Al ₂ O ₃ in Tank 241-AZ-101 Waste	128
6-3. Al ₂ O ₃ and Fe ₂ O ₃ Concentrations in Glass versus Al ₂ O ₃ in Tank 241-AZ-101 . . .	129
6-4. Change in Volume of Glass versus Percent Change of Al ₂ O ₃ in Tank 241-AZ-101 Waste	129
6-5. Effect of Liquidus Temperature, Nepheline, and Composition Variability Study Limits on Volume of Glass versus Fe ₂ O ₃ in Tank 241-AZ-101 Waste	130
6-6. Effect of High-Level Waste Glass Model and Composition Variability Study Limits on Volume of Glass versus Fe ₂ O ₃ in Tank 241-AZ-101 Waste	130

LIST OF FIGURES (CONTINUED)

6-7. Fe_2O_3 and Al_2O_3 Concentrations in Glass versus Fe_2O_3 in Tank 241-AZ-101 Waste	131
6-8. Change in Volume of Glass versus Percent Change of Fe_2O_3 in Tank 241-AZ-101 Waste	131
6-9. Effect of Liquidus Temperature, Nepheline, and Composition Variability Study Limits on Volume of Glass versus Al_2O_3 in Tank 241-C-106 Waste	135
6-10. Effect of High-Level Waste Glass Model and Composition Variability Study Limits on Volume of Glass versus Al_2O_3 in Tank 241-C-106 Waste	135
6-11. Al_2O_3 and Fe_2O_3 Concentrations in Glass versus Al_2O_3 in Tank 241-C-106 Waste	136
6-12. Change in Volume of Glass versus Percent Change of Al_2O_3 in Tank 241-C-106 Waste	136
6-13. Effect of Liquidus Temperature, Nepheline, and Composition Variability Study Limits on Volume of Glass versus Fe_2O_3 in Tank 241-C-106 Waste	137
6-14. Effect of High-Level Waste Glass Model and Composition Variability Study Limits on Volume of Glass versus Fe_2O_3 in Tank 241-C-106 Waste	137
6-15. Fe_2O_3 and Al_2O_3 Concentrations in Glass versus Fe_2O_3 in Tank 241-C-106 Waste	138
6-16. Change in Volume of Glass versus Percent Change of Fe_2O_3 in Tank 241-C-106 Waste	138

LIST OF TABLES

3-1. Tank 241-AZ-101 Waste Composition	13
3-2. Tank 241-AZ-102 Waste Composition.	14
3-3. Tank 241-AY-102 Waste Composition	16
3-4. Tank 241-C-106 Waste Composition.	18

LIST OF TABLES (CONTINUED)

3-5. Nominal Blend Composition	20
4-1. Liquid-Fed Ceramic Melter Decontamination Factors	30
4-2. Corrosion Rates for Candidate Electrode Materials in HTB-650 Glass at 1,400 °C	35
4-3. Glass Production Rates for the Small-Scale High-Temperature Melter	38
4-4. Glass Production Rates for Low Temperature Melters	39
4-5. Melter Decontamination Factors for High-Temperature and Low-Temperature Melter	41
4-6. Percent of Feed Components Volatilized During Low-Level Waste Melter Tests	43
4-7. Composition and Volume Percent Crystals in Small-Scale High Temperature Melter Tests 1 and 2	43
4-8. Principal Radionuclides in Savannah River's Waste	45
5-1. Waste Acceptance Preliminary Specifications for the Defense Waste Processing Facility High-Level Waste Form	51
5-2. Hanford Waste Vittrification Plant Glass Property Constraints	52
5-3. Glass Composition Range Covered By Composition Variability Study (CVS-1 and CVS-2) Composition Region	53
5-4. Fulcher First-Order Viscosity Model Coefficients A, B and T_i	55
5-5. Arrhenius Equation Electrical Conductivity Coefficients A and B	57
5-6. Component Coefficients for First-Order Model of Liquidus Temperature	59
5-7. First and Second-Order Spinel Liquidus Temperature Model Coefficients B, C and D	65
5-8. Comparison of Measured to Predicted Spinel Liquidus Temperatures for Phase I Glasses and Selected Group of CVS and Savannah River Glasses	66

LIST OF TABLES (CONTINUED)

5-9. Environmental Assessment Glass Dissolution Rates	68
5-10. Coefficients for Second-Order Model (#3) of Natural Logarithm of Average 7-Day Product Consistency Test Normalized Elemental Releases	77
5-11. Chloride and Fluoride Glass Compositions and Percent Retained During Pilot-Scale Ceramic Melter-22 and Pilot-Scale Ceramic Melter-23 Melter Tests	82
5-12. Noble Metals Feed Composition and Percent Retained in the PAMELA Melter . .	85
5-13. Noble Metals Feed Composition and Percent Retained in the Savannah River Site Test Melter	86
5-14. Noble Metals Feed Composition, Percent Retained, and Average Glass Composition in Large-Scale Japanese Melter Tests	87
5-15. Minor Component Glass Solubility Limits for High-Level Waste	89
5-16. Comparison of Radioactive Sample and Simulant Rheology Properties	91
5-17. Durability Results for 241-AZ-101, Core 1 Glass	92
5-18. Durability Results for 241-AZ-101, Core 2 Glass	92
5-19. Durability Results for 241-AZ-102, Core 1 Glass	92
6-1. Waste and Blend Compositions in Mass Fractions on Nonvolatile Oxide Basis	97
6-2. Waste Loadings, Glass Compositions, Glass Property Predictions, and Additive Composition: Nominal Blend Waste	101
6-3. Waste Loadings, Glass Compositions, Glass Property Predictions, and Additive Composition: AZ Blend Waste	102
6-4. Waste Loadings, Glass Compositions, Glass Property Predictions, and Additive Composition: 241-C-106 Tank Waste	103
6-5. Waste Loading, Measured Glass Properties and Crystalline Phases in the Water Washed and Caustic Washed Nominal Blend Glasses	105

LIST OF TABLES (CONTINUED)

6-6. Waste Loading, Measured Property Values, and Crystalline Phases in Water Washed AZ Blend Glasses	106
6-7. Waste Loading, Measured Glass Properties and Crystalline Phases in the Water Washed 241-C-106A and 241-C-106B Glasses	107
6-8. Development of Bounding Composition Limits for the (Water Washed) Nominal Blend Glass	112
6-9. Development of Bounding Composition Limits for the (Caustic Washed) Nominal Blend Glass	114
6-10. Development of Bounding Composition Limits for the AZ Blend Glass	116
6-11. Development of Bounding Composition Limits for the 241-C-106A Glass	118
6-12. Development of Bounding Composition Limits for the 241-C-106B Glass	120
6-13. Properties of the Phase I Experimental Glasses	121
6-14. Glass Volume Estimates for Phase I Experimental Glasses	122
6-15. Glass Volume Estimates for Unblended Phase I Wastes	123
6-16. Glass Volume Estimates for Various Phase I Blends	124
6-17. Glass Volume as a Function of the Amount of Al_2O_3 in Tank 241-AZ-101 Waste	126
6-18. Glass Volume as a Function of the Amount of Fe_2O_3 in Tank 241-AZ-101 Waste	127
6-19. Glass Volume as a Function of the Amount of Al_2O_3 in Tank C-106 Waste	134
6-20. Glass Volume as a Function of the Amount of Fe_2O_3 in Tank 241-C-106 Waste	134
6-21. Tank 241-AZ-101 Base Case Glass Composition	140

LIST OF TABLES (CONTINUED)

6-22. Waste Concentration Change Needed to Induce a 50 °C Change in Liquidus Temperature (T_L) or 5 Percent Change in the Nepheline Limit for Tank 241-AZ-101 Glass	141
6-23. Tank 241-C-106 Base Case Glass Composition	141
6-24. Waste Concentration Change Needed to Induce a 50 °C Change in Liquidus Temperature (T_L) or 5 Percent Change in the Nepheline Limit for Tank 241-C-106 Glass	142
6-25. Chemical Composition of Offgas Deposits	146
6-26. Plutonium Inventories in Tanks 241-AZ-101, 241-AZ-102, 241-C-106, and 241-AY-102	148
6-27. Maximum Plutonium Oxide Concentration in Phase I Melter Feeds	149
7-1. Glass Volume Estimates for Phase I Experimental Glasses	157
7-2. Glass Volume Estimates for Unblended Phase I Wastes	158
7-3. Glass Volume Estimates for Various Phase I Blends	159
7-4. Glass Volume as a Function of the Amount of Al_2O_3 in Tank 241-AZ-101 Waste	161
7-5. Glass Volume as a Function of the Amount of Al_2O_3 in Tank 241-C-106 Waste	163
7-6. Waste Concentration Change Needed to Induce a 50 °C Change in Liquidus Temperature (T_L) or 5 Percent Change in the Nepheline Limit for Tank 241-AZ-101 Glass	164

LIST OF ABBREVIATIONS AND ACRONYMS

AFO	Adjusted feed oxide
BACT	Best available control technology
CC	Complexant concentrate
CCC	Canister centerline cooling
CERCLA	Comprehensive Environmental Response, Compensation, and Liability Act
CVS	Composition variability study
CW	Caustic washed
CWP	PUREX Coating Waste
DF	Decontamination factor
DOE	U.S. Department of Energy
DOE-OCRWM	U.S. Department of Energy-Office of Environmental Restoration and Waste Management
DRE	Destruction and removal efficiency
DSSF	Double-shell slurry feed
DST	Double-shell tank
DWPF	Defense Waste Processing Facility
EA	Environmental assessment
Ecology	Washington State Department of Ecology
EDTA	Ethylenediametetraacetic acid
EIS	Environmental Impact Statement
FMEA	Failure modes and effects analysis
FY	Fiscal year
HEDTA	N-(hydroxyethyl)-ethylenediaminetriacetic acid
HLW	High-level waste
HTM	High-temperature melter
HWVP	Hanford Waste Vittrification Plant
IDMS	Integrated DWPF Melter System
IHLW	Immobilized high-level waste
Kfk	Kernforschungszentrum Karlsruhe
LANL	Los Alamos National Laboratory
LAW	Low-Activity Waste
LFCM	Liquid-fed ceramic melter
LLW	Low-level waste
LTM	Low-temperature melter
MCC	Materials Characterization Center
MT	Metric Tons
NBO	Non-bridging oxygen
NCAW	Neutralized current acid waste
NCRW	Neutralized cladding removal waste
ORIGEN	Oak Ridge Isotope Generation (computer code)
PCCS	Product composition control system
PCT	Product consistency test

LIST OF ABBREVIATIONS AND ACRONYMS (CONTINUED)

PFP	Plutonium Finishing Plant
PNNL	Pacific Northwest National Laboratory
PSCM	Pilot-scale ceramic melter
PUREX	Plutonium-Uranium Extraction
QCM	Quasichemical model
RCRA	Resource Conservation and Recovery Act of 1976
REDOX	Reduction oxidation
RFP	Request for Proposal
RLFCM	Radioactive, liquid-fed ceramic melter
RSM	Research scale melter
SCF	Shielded cells facility
SCR	Selective catalytic reduction
SEM/EDS	Scanning electron microscopy/energy dispersive spectroscopy
SGM	Scale glass melter
SME	Slurry mix evaporator
SRS	Strontium removal sludge
SRS	Savannah River Site
SRTC	Savannah River Technology Center
SS	Stainless steel
SSHTM	Small-scale high temperature melter
SST	Single-shell tank
TEM	Transmission electron microscopy
TLM	Tank-layer model
TOC	Total Organic Carbon
TOE	Total online efficiency
Tri-Party Agreement	<i>Hanford Federal Facility Agreement and Consent Order</i>
TRU	Transuranic
TWRS	Tank Waste Remediation System
UR	Uranium Recovery
USBM	U.S. Bureau of Mines
WAPS	Waste acceptance preliminary specifications
WCP	Waste form compliance plan
WIPP	Waste Isolation Pilot Plant
WOL	Waste oxide loading
WQR	Waste for qualification report
WSTC	Westinghouse Science and Technology Center
WSTRS	Waste Status and Transaction Record Summaries
WVDP	West Valley Demonstration Project
WW	Water washed
XRD	X-ray diffraction

TANK WASTE REMEDIATION SYSTEM PHASE I HIGH-LEVEL WASTE FEED PROCESSABILITY ASSESSMENT REPORT

1.0 INTRODUCTION

1.1 BACKGROUND

The primary mission of the high-level waste (HLW) program is to process and immobilize the Hanford Site high-level defense waste into a stable borosilicate glass product (or acceptable nonborosilicate waste form) for the Federal geologic repository.

The U.S. Department of Energy (DOE) has determined that this mission will be accomplished by acquiring Hanford Site tank waste treatment and immobilization services using privatized facilities. These facilities will be privately developed, owned, and operated under the requirements of the *Resource Conservation and Recovery Act of 1976* (RCRA). Moreover, DOE, the U.S. Environmental Protection Agency, and the Washington State Department of Ecology (Ecology) have entered into the *Hanford Federal Facility Agreement and Consent Order*, referred to as the Tri-Party Agreement, to ensure compliance with RCRA and the *Comprehensive Environmental Response, Compensation, and Liability Act* (CERCLA). The Tri-Party Agreement sets forth an aggressive schedule for starting hot operations of the low-activity waste (LAW) pretreatment facility (December 2004) and LAW vitrification facility (June 2005); and commencing hot operations of the HLW pretreatment facility (June 2008) and HLW immobilization facility (December 2009).

It is anticipated that the Hanford Site tank waste disposal mission will be completed in two phases: a Phase I demonstration where 3 to 6 percent of the HLW will be processed at demonstration scale over a period of 5 to 9 years, and a Phase II program where commercial scale facilities will be used to treat the remaining inventory of the Hanford Site's tank waste. The objectives of this initial phase are to demonstrate the commercial feasibility of using private facilities to treat and immobilize the Hanford Site tank waste. After the completion of this initial phase, DOE plans to competitively award contracts for production scale facilities. The Request for Proposal (RFP) for Phase I provides the schedule for the Phase I contract award date (December 29, 1997), completion of design (December 31, 1999), starting hot operations of the HLW immobilization facility (June 1, 2007), processing the minimum quantity of HLW (June 1, 2011), and decontamination and decommissioning and RCRA closure (June 1, 2013).

Most of the HLW technology issues must be resolved before completion of detailed design for Phase I. The objective of this study is to evaluate the composition of the feeds and to analyze the properties of vitrified waste for Phase I HLW privatization. This information is necessary for development of the feed and product (waste form) specifications for the Phase I RFP, to assess the need for caustic washing, to improve estimates of Phase I

interim storage requirements (quantity of glass and waste oxide loading) and to provide an acceptable basis for independently evaluating vendor estimates for Phase I glasses (if a glass waste form is proposed). Phase I waste compositions are expected to have a significant affect on the operating performance of a glass melter (melter processing capacity and sludge accumulation potential) and performance properties of the glass.

The volume and composition of waste in the single-shell tanks (SSTs) and double-shell tanks (DSTs) vary significantly (Hanlon 1994). Before 1992, the Hanford Waste Vitrification Plant (HWVP) scope was limited to processing only DST waste, which includes neutralized current acid waste (NCAW), neutralized cladding removal waste (NCRW), Plutonium Finishing Plant (PFP) waste, and complexant concentrate (CC) waste. These wastes contain high concentrations of chromium, zirconium, and noble metals (Ag, Pd, Rh, Te, and Ru) that may require special pretreatment or blending to avoid possible sludge accumulation problems in the melter. With the addition of SST wastes, the waste composition envelope has expanded to include other wastes known to be high in aluminum, phosphate, chromium, and uranium, that may limit the acceptable glass composition range and increase the volume of HLW glass produced. Such wastes must be thoroughly characterized to ensure that the vitrification process will produce glasses (or alternate waste forms) of acceptable quality for the waste repository.

In-tank sludge washing and caustic leaching processes can be used to reduce the level of certain non-radioactive and glass limiting components in the waste. These processes are especially needed to reduce the amount of sodium, aluminum, chrome, sulfur, and phosphorus in the HLW melter feed and to add these components to the low activity stream for LAW vitrification. Laboratory studies have indicated that these processes are likely to be effective for certain wastes and will reduce the volume of HLW glass and improve performance characteristics of the HLW melter.

Better estimates of the pretreated waste and melter feed compositions are needed to support the design and operation of the HLW immobilization facility. Compositions derived from tank waste composition models, characterization data and process flowsheet, and simulation models of the pretreatment and vitrification processes will be evaluated to ensure that the pretreated waste can be processed, as required, into a qualified glass product (or acceptable alternate waste form).

Since the alternate waste forms have not been identified, this study will be limited to borosilicate glass as the preferred waste form for the Phase I HLW.

1.2 PURPOSE

The purpose of this report is to provide vitrification basis for pretreatment and blending of Phase I HLW and to evaluate the impact of feed composition on operation of the HLW immobilization facility. This includes developing estimates of total HLW glass volume based

on limiting components in the waste using different scenarios for tank waste blending and pretreatment, and on glass composition ranges and limits.

1.3 SCOPE

This assessment focuses on estimating the composition of wastes for Phase I HLW immobilization, processing those wastes through the pretreatment and vitrification functions, evaluating melter performance characteristics, and estimating the volume and composition of the HLW glasses. Estimates of the waste oxide loading (WOL) and glass composition will be determined for Phase I wastes and blends of those wastes. Potential benefits of sludge washing and caustic leaching will be assessed to determine the likely impact on glass volume. Each HLW glass will be evaluated for acceptable viscosity, electrical conductivity and absence of crystals at the melter operating temperature, and acceptable leach resistance after disposal. Glass volume predictions will be generated based on the limiting properties of the glass, including liquidus temperature and nepheline constraints. This assessment will address balance of plant issues, considering corrosion of offgas system components, volatile emissions, and offgas treatment.

This document will be revised periodically as new information on the properties and composition of Phase I wastes becomes available. Previous assessments, including the *Hanford Waste Vitrification Plant Melter Feed Criteria Recommendation* (Stegen and Baker 1990) and *Hanford Waste Vitrification Plant Feed Processability Assessment* (May and Watrous 1991), evaluated the acceptability of estimated HWVP (DST) feed compositions and identified those areas where feed composition would have an adverse affect on the operability of the HWVP. The scope of this assessment includes only those wastes selected for Phase I HLW immobilization and focuses on the integrated performance of the retrieval and pretreatment functions and HLW immobilization process. Tank characterization data will be used to construct estimates of the average and bounding melter feed compositions and related glass composition limits for the HLW plant. Later revisions to this document will discuss the planning and processing impacts of waste retrieval, blending, and sludge pretreatment processes for Phase II HLW feeds and how these processes should be integrated into the Hanford Site flowsheet (Orme 1995).

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2.0 PHASE I HIGH-LEVEL WASTES

The HLW immobilization facility will receive waste under conditions described in the RFP and will process those wastes to make a vitrified glass product. This section describes: (1) the selection of candidate Phase I wastes based on waste consolidation plans, (2) the origin and principal characteristics of this waste, and (3) estimated compositions based on sludge samples from the candidate tanks. Tank waste compositions developed from this assessment will be used to derive composition estimates for water and caustic washed sludges and to develop melter feed composition estimates for various feed batches or waste blends in Section 3.0. These compositions will also be used to determine the composition and properties of the corresponding HLW glasses in Section 6.0.

2.1 SELECTION OF PHASE I WASTES

The candidate HLW feeds for Phase I are NCAW in Tanks 241-AZ-101 and 241-AZ-102, and high-heat sludge retrieved from Tank 241-C-106 into Tank 241-AY-102 (Bacon 1996). Prewashed solids being prepared for HLW processing will be stored in three tanks (241-AZ-101, 241-AZ-102, and 241-AY-102). It is unlikely that any of these wastes will be mixed until the tank safety issues have been resolved, including: (1) heat and sludge height removal limits for Tanks 241-AZ-101 and 241-AZ-102, (2) criticality limits for Tanks 241-AZ-101 and 241-AZ-102, (3) the five molar sodium concentration limit for aging tank waste supernatants, and (4) the radionuclide source term limits for 242-A evaporator processing of supernates.

Sludges in Tanks 241-AZ-101, 241-AZ-102 and 241-AY-102 may be subsequently leached and washed, if necessary, before transfer of these wastes to the HLW immobilization facility. However, tank space management and conservation needs, and tank safety risk considerations may limit the possible use of caustic leaching processes for these waste (Powell 1995).

2.2 SOURCE OF WASTES

Tank 241-AZ-101 wastes primarily consist of NCAW from the last Plutonium-Uranium Extraction (PUREX) production campaign, that were transferred into Tank 241-AZ-101 from 1983 through March 13, 1986. Before this transfer, approximately 21,500 L of sludge remained as a residual heel in this tank. The current inventory in Tank 241-AZ-101 is approximately 130,000 L of sludge (Hodgson 1995). Tank 241-AZ-102 waste is similar in most respects, consisting of NCAW waste transferred to Tank 241-AZ-102 from March 13, 1986, through the end of the last PUREX production campaign in 1989. Before the initial transfers of NCAW waste, Tank 241-AZ-102 contained about 31,500 L of sludge from other unspecified sources. The current inventory in Tank 241-AZ-102 is approximately 333,000 L of sludge (Ryan 1995).

Tank 241-AY-102 wastes mainly consist of settled solids from dilute supernate transfers (66,000,000 to 68,000,000 L), aging waste transfers from 1977 to 1978, double-shell slurry feed (DSSF) transfers in 1980, and dilute, non-complexed waste after 1980. The original sources include B Plant, PUREX, SSTs, 100 Area, 300 Area, 400 Area, T Plant, S Plant, and various DSTs. The sludge volume was initially determined to be 23,000 L on September 14, 1978. Approximately 2,700 L of this total may have consisted of sodium aluminate rich sludge from DSSF, while the remaining sludge was probably from NCAW and commercial vitrification process waste. The sludge inventory increased to 80,000 L by June 30, 1980, including 57,000 L of sludge from DSSF waste.

Currently, Tank 241-AY-102 contains 121,000 L of sludge from various unidentified sources. The chemical composition of this sludge, including the ratio of aluminum to iron, is consistent with PUREX type sludges from B Plant, while the high silver and chloride inventories in this waste (559 kg and 644 kg, respectively) suggests the possible receipt of silver catalyst from the B Plant cesium and strontium purification operations of Isochem (involving less than 1,270 kg Ag) (Buckingham 1967).

The LANL Tank Layer Model (TLM) can be used as a basis for estimating the sludge layers in Tank 241-C-106 (Agnew 1995). According to this model, Tank 241-C-106 waste mainly consists of: (1) uranium recovery process sludge (UR waste) (102,000 L), (2) PUREX coating waste (CWP) (128,700 L), (3) washed PUREX sludge from B Plant cesium/strontium recovery operations (AR vault waste) (242,200 L), (4) LLW from B Plant operations (BL waste) (75,500 L), and (5) unknown waste (probably AR and BL waste) from B Plant (196,800 L) (Agnew 1995).

2.3 GENERAL CHARACTERISTICS OF WASTES

Approximately 1,877 ML of waste have been produced at the Hanford Site by the reprocessing of irradiated fuel from the plutonium production reactors. The principal chemical separation processes include the bismuth phosphate process (T and B Plants, 1944 to 1956), the reduction oxidation (REDOX) process (1952 to 1966), and the PUREX solvent extraction process (1956 to 1972, 1983 to 1989). Certain tank wastes, such as the metal waste stream from the bismuth phosphate process, were reworked in U Plant to recover uranium (uranium recovery process, 1952 to 1957). During the same period, low-level supernate wastes were treated to remove ^{137}Cs and ^{90}Sr before disposing of such wastes to cribs or trenches (nickel ferrocyanide scavenging operations disposing of 325 ML of waste). From 1965 to 1976, high-heat PUREX waste sludges and general supernate wastes were reworked in B Plant to recover ^{137}Cs and ^{90}Sr by ion exchange and solvent extraction, respectively (cesium and strontium encapsulation operation).

Because of severe limitations in tank space, some wastes were concentrated by in-tank evaporation or external evaporation processes (242-B and T evaporators, in-tank BY-ITS, S, and SX farm evaporators and 242-A and S evaporator-crystallizers) with condensates being routed to cribs and trenches, recycled to other tanks or treated for subsequent disposal. Approximately 1,139 ML of waste water were eventually eliminated by various evaporation processes. By employing such processes, wastes in SST farms were gradually converted into mixed sludges and salt cake, while most of the recent wastes from PUREX and supernates from older wastes have been consolidated into the double-shell tank farms. Approximately 174 ML of waste remain in the tanks from the 1944 to 1980 era, together with 57 ML of concentrated waste from recent operations at the Hanford Site...

Figures 2-1 and 2-2 show the composition of the principal streams from the uranium recovery process (UR sludge) and from the PUREX and B Plant processes (PUREX coating waste, neutralized PUREX HLW, and washed PUREX sludge from B Plant) (GE 1951 and 1955, RHO 1980). These waste compositions may be used with the tank transaction records to identify the tanks that received the wastes and as a basis for estimating the possible composition of this waste.

In-tank precipitation and mineralization processes have also changed the solubility properties of some components in the waste, such as sodium, aluminum, and NO_3 in cancrinite ($2 \text{ NaAlSiO}_4 \cdot 0.52 \text{ NaNO}_3 \cdot 0.68 \text{ H}_2\text{O}$), aluminosilicate and iron bismuth silicate hydroxide. Such wastes are often highly stratified with lateral and vertical inhomogeneities caused by the mixing and settling characteristics of the sludge, that makes it especially difficult in many cases to obtain a representative sample of the waste.

Figure 2-1. Composition of the Principal Streams from the Uranium Recovery Process.

Uranium recovery process

M = Molar concentration
 TBP = Tributyl phosphate
 UST = Underground storage tank

Figure 2-2. Composition of the Principal Streams from PUREX and B Plant Processes.

* Anderson, J.D., 1990, *A History of the 200 Area Tank Farms*, WHC-MR-0132, Westinghouse Hanford Company, Richland, Washington.

** Allen, G. K., 1976, *Estimated Inventory of Chemicals Added to Underground Waste Tanks, 1944 Through 1975*, ARH-CD-610 B, Atlantic Richfield Hanford Company, Richland, Washington.

M = Molar concentration
 PUREX = Plutonium-Uranium Extraction
 SRS = Strontium removal sludge
 TBP = Tributyl phosphate
 UST = Underground storage tank

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3.0 CURRENT ESTIMATES OF MELTER FEED COMPOSITION

This section provides estimates of the HLW melter feed composition based on available data. Estimates provided in this section are based on analysis of core samples and water washing tests performed with these samples, essential materials purchase records and for some components, flowsheets and fission yields keyed to the estimated ruthenium inventory (Hogan 1994, Enghusen 1987, Schofield 1989, 1990). These sources were used to estimate the composition of the sludge in Tanks 241-AZ-101, 241-AZ-102, 241-C-106, and 241-AY-102. At least one full depth core sample has been taken from each of these tanks. Core sample analysis reports and sludge level measurements were primarily used to determine the composition of most components in the Phase I wastes. This information is currently available from the Tank Characterization Reports for 241-AZ-101, 241-AZ-102, and 241-AY-102, and from the Data Transmittal package for 241-C-106 (Hodgson 1995, Ryan 1995, Gray 1993a, 1993b, Weiss 1987, McCown 1988, Hara 1990, Horton 1977, Bratzel 1980). The current inventory estimates for Tank 241-AZ-101 are based on the physical and chemical properties of the cores and average sludge level when the samples were taken in May 1989. The amount of aluminum in the waste of Tank 241-AZ-101 was estimated from essential materials purchase records and the analysis of supernate samples from this tank. It is anticipated that all Phase I wastes will be extensively water washed to remove most of the water soluble components, and for some wastes, caustic washing processes may be employed to reduce the amount of aluminum in these wastes. The melter feed composition estimates provided in this section assume that 88 percent of the water soluble components have been removed from the sludge during the waste retrieval and sludge washing operations in the tank farm.

3.1 241-AZ-101 TANK WASTE

Approximately 20 samples of waste from Tank 241-AZ-101 have been obtained since 1983, including several grab samples of liquid and sludge in 1987 and 1995. Tank 241-AZ-101 also was sampled in April and May of 1989, using a push-mode sampler to retrieve three core samples of sludge. About 200 g of sludge were recovered in the first core. Based on the physical properties of the core (density and solids weight), the first sludge layer appeared to be 23.2 cm (9.14 in.) thick at riser 15F where the first core sample was taken. About 513 g of sludge were obtained from the second core, corresponding to a sludge layer thickness of approximately 61 cm (24 in.) at riser 24D where this sample was taken. This value is in good agreement with the 58.4 cm (23 in.) sludge level that was measured from riser 24D several months before the core sample was taken. If the sludge depth corresponding to the core sample data from riser 24D is included with data from the other risers in calculating sludge depth, the average sludge level is 39.2 cm (15.43 in.), corresponding to a sludge inventory of 161,000 L. This figure was used to estimate component inventories in Tank 241-AZ-101. During the analysis of the core samples, sludge and liquid from the core were composited and subjected to water washing tests using a

gravity settling/liquid decant procedure. The tank waste composition estimates for this waste are provided in Table 3-1.

3.2 241-AZ-102 TANK WASTE

Several grab samples of liquid and sludge were obtained in 1987, and a two segment core of the sludge in May and July 1989, from Tank 241-AZ-102. Other dip samples were collected as recently as February 1995. The 1989 core sample was taken from riser 15L and contained 268 g of solids and 25 g of free draining liquid in the first segment and 287 g of solids and negligible liquid in the second segment. Based on the physical properties and appearance of the core, the sludge layer appeared to be 85.4 cm (33.6 in.) thick at the location of the core sample. This estimate is consistent with the sludge level measurements taken on May 30, 1989. The average layer was 81.12 cm (31.94 in.) thick on May 30, 1989. Based on these results, the inventory at the time of sampling was 333,000 L of sludge (Hanlon 1993). The sludge and liquid samples from this core were composited and subjected to a similar series of water washing tests. The tank waste composition estimates are provided in Table 3-2.

3.3 241-AY-102 TANK WASTE

A core sample of sludge was taken in 1987 and grab samples of liquid in June and December 1994 from Tank 241-AY-102. The core sample, taken from riser 13A, consisted of two segments that were divided into four subsamples, three of which contain solids and the other supernate and about one percent solids from the top segment of the core. The sludge volume at the time of the 1987 core sample was 121,000 L (32,000 gal). This volume is consistent with the current sludge inventory in Tank 241-AY-102 (Hanlon 1993). The estimated composition of this waste is provided in Table 3-3.

3.4 241-C-106 TANK WASTE

Several samples of liquid and sludge were obtained from Tank 241-C-106 in 1977 and 1980, and a four segment core sample of sludge from riser 1 in 1986. About 1,300 g of sludge and 80 ml of liquid were recovered in the 1986 core sample. Based on the physical properties of the core, the sludge layer appeared to be 193 cm (76 in.) thick at the sampling location. This sludge depth is consistent with sludge level measurements that, at the time of sampling, indicate the sludge layer was 196.3 cm (77.3 in.) thick. The current inventory is about 745,000 L corresponding to an average sludge depth of 181.6 cm (71.5 in.) (Hanlon 1993). A composite sample of 239.5 g of sludge was formed from the core material. Water washing and acid leaching tests (with HNO_3 and HNO_3 -HF-HCl) were performed using a centrifuge separation/liquid decant procedure. The tank waste composition estimates for this waste are provided in Table 3-4.

WHC-SD-WM-TI-768
Revision 0

Table 3-1. Tank 241-AZ-101 Waste Composition.

241-AZ-101 Element	Element wt kg	241-AZ-101 Oxide	Oxide factor	Oxide wt kg	Oxide wt%	Water wash factor	Water wash solids, kg	Water wash solid, wt%	Caustic wash factor	Caustic wash solids, kg	Caustic wash solids, wt%
Al	13384.00	Al2O3	1.89	25282.38	20.74	1.00	25282.38	23.15	0.52	13146.8	13.69
As	136.30	As2O3	1.32	179.92	0.15	1.00	179.92	0.16	1.00	179.92	0.19
Ba	174.90	BaO	1.12	195.19	0.16	1.00	195.19	0.18	1.00	195.19	0.20
Be	4.21	BeO	2.78	11.68	0.01	1.00	11.68	0.01	1.00	11.68	0.01
B	71.80	B2O3	3.21	230.26	0.19	1.00	230.26	0.21	1.00	230.26	0.24
Ca	587.00	CaO	1.40	821.21	0.67	1.00	821.21	0.75	1.00	821.21	0.86
Cd	1357.35	CdO	1.14	1550.09	1.27	1.00	1550.09	1.42	1.00	1550.09	1.61
Ce	293.27	CeO2	1.23	360.14	0.30	1.00	360.14	0.33	1.00	360.14	0.38
Cl	26.75	Cl-	1.00	26.75	0.02	0.74	19.80	0.02	1.00	19.80	0.02
Cr	398.00	Cr2O3	1.46	581.48	0.48	0.70	407.03	0.37	0.89	362.26	0.38
Co	279.89	Co2O3	1.41	393.81	0.32	1.00	393.81	0.36	1.00	393.81	0.41
Cu	104.65	CuO	1.25	131.02	0.11	1.00	131.02	0.12	1.00	131.02	0.14
Dy	20.43	Dy2O3	1.15	23.43	0.02	1.00	23.43	0.02	1.00	23.43	0.02
F	544.67	F-	1.00	544.67	0.45	0.41	223.31	0.20	1.00	223.31	0.23
Fe	24086.84	Fe2O3	1.43	34444.18	28.26	1.00	34444.18	31.54	1.00	34444.1	35.87
La	910.76	La2O3	1.17	1068.32	0.88	1.00	1068.32	0.98	1.00	1068.32	1.11
Pb	127.70	PbO	1.08	137.53	0.11	1.00	137.53	0.13	1.00	137.53	0.14
Li	17.59	Li2O	2.15	37.85	0.03	1.00	37.85	0.03	1.00	37.85	0.04
Mg	147.25	MgO	1.66	244.14	0.20	1.00	244.14	0.22	1.00	244.14	0.25
Mn	728.34	MnO2	1.58	1152.23	0.95	1.00	1152.23	1.06	1.00	1152.23	1.20
Mo	26.75	MoO3	1.50	40.13	0.03	1.00	40.13	0.04	1.00	40.13	0.04
Nd	651.43	Nd2O3	1.17	760.22	0.62	1.00	760.22	0.70	1.00	760.22	0.79
Ni	1073.75	NiO	1.27	1366.88	1.12	1.00	1366.88	1.25	1.00	1366.88	1.42
P	674.47	P2O5	2.29	1545.21	1.27	1.00	1545.21	1.41	0.34	525.37	0.55
Pu	14.87	PuO2	1.13	16.82	0.01	1.00	16.82	0.02	1.00	16.82	0.02
K	1420.52	K2O	1.21	1711.73	1.40	1.00	1711.73	1.57	1.00	1711.73	1.78
Re	13.87	Re2O3	1.13	15.66	0.01	1.00	15.66	0.01	1.00	15.66	0.02
Rh	105.15	Rh2O3	1.23	129.65	0.11	1.00	129.65	0.12	1.00	129.65	0.14
Ru	214.00	Ru2O3	1.23	263.22	0.22	1.00	263.22	0.24	1.00	263.22	0.27
Sb	656.38	Sb2O3	1.20	785.69	0.64	1.00	785.69	0.72	1.00	785.69	0.82
Se	432.97	SeO2	1.41	608.32	0.50	1.00	608.32	0.56	1.00	608.32	0.63
Si	1410.61	SiO2	2.14	3017.29	2.48	1.00	3017.29	2.76	1.00	3017.29	3.14
Ag	98.95	Ag2O	1.07	106.27	0.09	1.00	106.27	0.10	1.00	106.27	0.11
Na	18058.00	Na2O	1.35	24342.18	19.97	0.60	14605.31	13.37	1.00	14605.3	15.21
Sr	119.88	SrO	1.18	141.82	0.12	1.00	141.82	0.13	1.00	141.82	0.15
SO4	3485.00	SO4	1.00	3485.00	2.86	0.30	1045.50	0.96	1.00	1045.50	1.09
Te	464.67	TeO2	1.25	582.23	0.48	1.00	582.23	0.53	1.00	582.23	0.61
Th	299.46	ThO2	1.14	340.79	0.28	1.00	340.79	0.31	1.00	340.79	0.35
Ti	158.77	TiO2	1.67	264.83	0.22	1.00	264.83	0.24	1.00	264.83	0.28
Tl	1619.54	Tl2O3	1.12	1809.03	1.48	1.00	1809.03	1.66	1.00	1809.03	1.88
U	1328.37	U3O8	1.18	1566.15	1.28	1.00	1566.15	1.43	1.00	1566.15	1.63
V	6.19	V2O5	1.79	11.05	0.01	1.00	11.05	0.01	1.00	11.05	0.01
Zn	97.59	ZnO	1.24	121.40	0.10	1.00	121.40	0.11	1.00	121.40	0.13
Zr	8478.51	ZrO2	1.35	11445.99	9.39	1.00	11445.99	10.48	1.00	11445.9	11.92
Total Cation, kg	84311.40	Total Oxides		121893.8	100		109214.7	100		96014.5	100
Total Anions (NO2, NO3, TIC as CO3), kg	31283.50										
Water, kg	115090.00										
Total sludge kg (den = 1.66 kg/42.05 % water)	230684.90										
Total estimated sludge, kg	267260.00										
Mass balance percent	86.46										
Missing:	Ox.										

WHC-SD-WM-TI-768
Revision 0

Table 3-2. Tank 241-AZ-102 Waste Composition. (2 sheets)

Element	Element wt kg	Oxide	Oxide factor	Oxide wt kg	Oxide wt%	Water wash factor	Water wash solids, kg	Water wash solids, wt%	Caustic wash factor	Caustic wash solids, kg	Caustic wash solids, wt%
Al	19184.00	Al ₂ O ₃	1.89	36238.58	20.06	1.00	36238.58	20.95	0.52	18844.06	12.16
As	211.48	As ₂ O ₃	1.32	279.15	0.15	1.00	279.15	0.16	1.00	279.15	0.18
Ba	197.55	BaO	1.12	220.47	0.12	1.00	220.47	0.13	1.00	220.47	0.14
Be	4.80	BeO	2.78	13.32	0.01	1.00	13.32	0.01	1.00	13.32	0.01
B	177.54	B ₂ O ₃	3.21	569.37	0.32	1.00	569.37	0.33	1.00	569.37	0.37
Ca	1078.87	CaO	1.40	1509.34	0.84	1.00	1509.34	0.87	1.00	1509.34	0.97
Cd	5358.64	CdO	1.14	6119.57	3.39	1.00	6119.57	3.54	1.00	6119.57	3.95
Ce	280.43	CeO ₂	1.23	344.37	0.19	1.00	344.37	0.20	1.00	344.37	0.22
Cl	52.51	Cl-	1.00	52.51	0.03	0.60	31.51	0.02	1.00	31.51	0.02
Cr	685.90	Cr ₂ O ₃	1.46	1002.10	0.55	0.87	871.83	0.50	0.89	775.93	0.50
Co	525.14	Co ₂ O ₃	1.41	738.87	0.41	1.00	738.87	0.43	1.00	738.87	0.48
Cu	125.03	CuO	1.25	156.54	0.09	1.00	156.54	0.09	1.00	156.54	0.10
Dy	18.00	Dy ₂ O ₃	1.15	20.65	0.01	1.00	20.65	0.01	1.00	20.65	0.01
F	164.33	F-	1.00	164.33	0.09	0.73	119.96	0.07	1.00	119.96	0.08
Fe	46798.75	Fe ₂ O ₃	1.43	66922.21	37.05	1.00	66922.21	38.69	1.00	66922.21	43.18
La	1607.59	La ₂ O ₃	1.17	1885.70	1.04	1.00	1885.70	1.09	1.00	1885.70	1.22
Pb	392.97	PbO	1.08	423.23	0.23	1.00	423.23	0.24	1.00	423.23	0.27
Li	10.43	Li ₂ O ₃	2.15	22.45	0.01	1.00	22.45	0.01	1.00	22.45	0.01
Mg	339.38	MgO	1.66	562.69	0.31	1.00	562.69	0.33	1.00	562.69	0.36
Mn	1028.86	MnO ₂	1.58	1627.66	0.90	1.00	1627.66	0.94	1.00	1627.66	1.05
Mo	16.79	MoO ₃	1.50	25.19	0.01	1.00	25.19	0.01	1.00	25.19	0.02
Nd	1039.57	Nd ₂ O ₃	1.17	1213.18	0.67	1.00	1213.18	0.70	1.00	1213.18	0.78
Ni	3165.17	NiO	1.27	4029.26	2.23	1.00	4029.26	2.33	1.00	4029.26	2.60
P	349.03	P ₂ O ₃	2.29	799.63	0.44	1.00	799.63	0.46	0.34	271.87	0.18
Pu	21.29	PuO ₂	1.13	27.19	0.02	1.00	27.19	0.02	1.00	27.19	0.02
K	743.06	K ₂ O	1.21	895.39	0.50	1.00	895.39	0.52	1.00	895.39	0.58
Re	25.93	Re ₂ O ₃	1.13	29.27	0.02	1.00	29.27	0.02	1.00	29.27	0.02
Rh	150.75	Rh ₂ O ₃	1.23	185.87	0.10	1.00	185.87	0.11	1.00	185.87	0.12
Ru	65.02	Ru ₂ O ₃	1.23	79.97	0.04	1.00	79.97	0.05	1.00	79.97	0.05
Sb	1218.20	Sb ₂ O ₃	1.20	1458.19	0.81	1.00	1458.19	0.84	1.00	1458.19	0.94
Se	367.96	SeO ₂	1.41	516.98	0.29	1.00	516.98	0.30	1.00	516.98	0.33
Si	1600.46	SiO ₂	2.14	3423.38	1.90	1.00	3423.38	1.98	1.00	3423.38	2.21
Ag	240.78	Ag ₂ O	1.07	258.60	0.14	1.00	258.60	0.15	1.00	258.60	0.17
Na	21113.03	Na ₂ O	1.35	28460.36	15.76	0.79	22341.39	12.91	1.00	22341.39	14.42
Sr	116.81	SrO	1.18	138.19	0.08	1.00	138.19	0.08	1.00	138.19	0.09
SO ₄	2336.36	SO ₄	1.00	2336.36	1.29	0.44	1028.00	0.59	1.00	1028.00	0.66
Te	123.24	TeO ₂	1.25	154.42	0.09	1.00	154.42	0.09	1.00	154.42	0.10
Th	173.98	ThO ₂	1.14	197.99	0.11	1.00	197.99	0.11	1.00	197.99	0.13
Ti	27.72	TiO ₂	1.67	46.24	0.03	1.00	46.24	0.03	1.00	46.24	0.03
Tl	2993.69	Tl ₂ O ₃	1.12	3343.95	1.85	1.00	3343.95	1.93	1.00	3343.95	2.16
U	4501.25	U ₃ O ₈	1.18	5306.97	2.94	1.00	5306.97	3.07	1.00	5306.97	3.42
V	11.64	V ₂ O ₅	1.79	20.78	0.01	1.00	20.78	0.01	1.00	20.78	0.01
Zn	50.37	ZnO	1.24	62.66	0.03	1.00	62.66	0.04	1.00	62.66	0.04
Zr	6466.09	ZrO ₂	1.35	8729.22	4.83	1.00	8729.22	5.05	1.00	8729.22	5.63
Total cation, kg	125160.39	Total		180612.34	100.00		172989.35	100.00		154971.18	100.00
Total anions (NO ₂ , NO ₃ , TIC as CO ₃), kg	41662.00										
Water, kg	253047.00										

Table 3-2. Tank 241-AZ-102 Waste Composition. (2 sheets)

Element	Element wt kg	Oxide	Oxide factor	Oxide wt kg	Oxide wt%	Water wash factor	Water wash solids, kg	Water wash solids, wt%	Caustic wash factor	Caustic wash solids, kg	Caustic wash solids, wt%
Tot. Sludg. kg (den=1.49kg/l, 51 % water	419869.39										
Total estimated sludge, kg	496170.00										
Mass Bal. %	84.62										

Table 3-3. Tank 241-AY-102 Waste Composition. (2 sheets)

Element	Element wt kg	Oxide	Oxide factor	Oxide wt. kg	Oxide wt %	Water wash factor	Water wash solids, kg	Water wash solids, wt%	Caustic wash factor	Caustic wash solids, kg	Caustic wash solids, wt%
Al	2919.00	Al ₂ O ₃	1.89	5513.99	17.42	1.00	5513.99	17.42	0.52	2867.28	10.15
As	0.00	As ₂ O ₃	1.32	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Ba	153.72	BaO	1.12	171.55	0.54	1.00	171.55	0.54	1.00	171.55	0.61
Be	0.00	BeO	2.78	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
B	213.97	B ₂ O ₃	3.21	686.20	2.17	1.00	686.20	2.17	1.00	686.20	2.43
Ca	1081.50	CaO	1.40	1513.02	4.78	1.00	1513.02	4.78	1.00	1513.02	5.35
Cd	32.00	CdO	1.14	36.54	0.12	1.00	36.54	0.12	1.00	36.54	0.13
Ce	85.70	CeO ₂	1.23	105.24	0.33	1.00	105.24	0.33	1.00	105.24	0.37
Cl	644.23	Cl-	1.00	644.23	2.04	1.00	644.23	2.04	1.00	644.23	2.28
Cr	271.13	Cr ₂ O ₃	1.46	396.12	1.25	1.00	396.12	1.25	0.89	352.55	1.25
Co	0.00	Co ₂ O ₃	1.41	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Cu	0.00	CuO	1.25	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Dy	0.00	Dy ₂ O ₃	1.15	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
F	655.82	F-	1.00	655.82	2.07	1.00	655.82	2.07	1.00	655.82	2.32
Fe	6465.40	Fe ₂ O ₃	1.43	9245.52	29.21	1.00	9245.52	29.21	1.00	9245.52	32.72
La	305.90	La ₂ O ₃	1.17	358.82	1.13	1.00	358.82	1.13	1.00	358.82	1.27
Pb	0.00	PbO	1.08	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Li	0.00	Li ₂ O ₃	2.15	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Mg	525.20	MgO	1.66	870.78	2.75	1.00	870.78	2.75	1.00	870.78	3.08
Mn	678.20	MnO ₂	1.58	1072.91	3.39	1.00	1072.91	3.39	1.00	1072.91	3.80
Mo	10.60	MoO ₃	1.50	15.90	0.05	1.00	15.90	0.05	1.00	15.90	0.06
Nd	205.40	Nd ₂ O ₃	1.17	239.70	0.76	1.00	239.70	0.76	1.00	239.70	0.85
Ni	233.20	NiO	1.27	296.86	0.94	1.00	296.86	0.94	1.00	296.86	1.05
P	461.70	P ₂ O ₃	2.29	1057.75	3.34	1.00	1057.75	3.34	0.34	359.64	1.27
Pu	3.06	PuO ₂	1.13	86.87	0.27	1.00	86.87	0.27	1.00	86.87	0.31
K	134.40	K ₂ O	1.21	161.95	0.51	1.00	161.95	0.51	1.00	161.95	0.57
Re	0.00	Re ₂ O ₃	1.13	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Rh	0.00	Rh ₂ O ₃	1.23	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Ru	0.00	Ru ₂ O ₃	1.23	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Sb	0.00	Sb ₂ O ₃	1.20	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Se	0.00	SeO ₂	1.41	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Si	934.68	SiO ₂	2.14	1999.28	6.32	1.00	1999.28	6.32	1.00	1999.28	7.08
Ag	559.20	Ag ₂ O	1.07	600.58	1.90	1.00	600.58	1.90	1.00	600.58	2.13
Na	3198.00	Na ₂ O	1.35	4310.90	13.62	1.00	4310.90	13.62	1.00	4310.90	15.26
Sr	12.23	SrO	1.18	14.47	0.05	1.00	14.47	0.05	1.00	14.47	0.05
SO ₄	47.50	SO ₄	1.00	47.50	0.15	1.00	47.50	0.15	1.00	47.50	0.17
Te	38.70	TeO ₂	1.25	48.49	0.15	1.00	48.49	0.15	1.00	48.49	0.17
Th	0.00	ThO ₂	1.14	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Ti	26.40	TiO ₂	1.67	44.04	0.14	1.00	44.04	0.14	1.00	44.04	0.16

Table 3-3. Tank 241-A-Y-102 Waste Composition. (2 sheets)

[illegible]

WHC-SD-WM-TI-768
Revision 0

Table 3-4. Tank 241-C-106 Waste Composition. (2 sheets)

Element	Element wt kg	Oxide	Oxide factor	Oxide wt kg	Oxide wt%	Water wash factor	Water wash solids, kg	Water wash solids, wt%	Caustic wash factor	Caustic wash solids, kg	Caustic wash solids, wt%
Al	43633.50	Al ₂ O ₃	1.89	82423.68	14.87	1.00	82382.47	16.97	0.52	42838.88	9.70
As	0.00	As ₂ O ₃	1.32	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Ba	5209.70	BaO	1.12	5814.03	1.05	1.00	5810.54	1.20	1.00	5810.54	1.32
Bi	533.90	Bi ₂ O ₃	1.12	595.30	0.11	0.97	-577.44	0.12	1.00	-577.44	0.13
B	20.80	B ₂ O ₃	3.21	66.71	0.01	0.76	50.39	0.01	1.00	50.39	0.01
Ca	12714.90	CaO	1.40	17788.15	3.21	1.00	17720.55	3.65	1.00	17720.55	4.01
Cd	394.50	CdO	1.14	450.52	0.08	0.97	435.92	0.09	1.00	435.92	0.10
Ce	0.00	CeO ₂	1.23	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Cl	0.00	Cl-	1.00	0.00	0.00	0.60	0.00	0.00	1.00	0.00	0.00
Cr	1049.70	Cr ₂ O ₃	1.46	1533.61	0.28	1.00	1531.46	0.32	0.89	1363.00	0.31
Co	5.10	Co ₂ O ₃	1.41	7.18	0.00	1.00	7.18	0.00	1.00	7.18	0.00
Cu	136.50	CuO	1.25	170.90	0.03	0.99	168.86	0.03	1.00	168.86	0.04
Dy	0.00	Dy ₂ O ₃	1.15	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
F	492.62	F-	1.00	492.62	0.09	0.73	359.61	0.07	1.00	359.61	0.08
Fe	55511.30	Fe ₂ O ₃	1.43	79381.16	14.33	1.00	79381.16	16.35	1.00	79381.16	17.97
La	0.00	La ₂ O ₃	1.17	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Pb	2564.30	PbO	1.08	2761.75	0.50	0.98	2710.11	0.56	1.00	2710.11	0.61
Li	0.00	Li ₂ O ₃	2.15	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Mg	6994.10	MgO	1.66	11596.22	2.09	1.00	11596.22	2.39	1.00	11596.22	2.62
Mn	1962.90	MnO ₂	1.58	3105.31	0.56	1.00	3105.31	0.64	1.00	3105.31	0.70
Mo	0.00	MoO ₃	1.50	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Nd (Hara)	283.62	Nd ₂ O ₃	1.17	330.98	0.06	1.00	330.98	0.07	1.00	330.98	0.07
Ni	1037.70	NiO	1.27	1320.99	0.24	0.97	1277.27	0.26	1.00	1277.27	0.29
P	3102.85	P ₂ O ₃	2.29	7108.63	1.28	0.85	6033.80	1.24	0.34	2051.49	0.46
Pu	88.22	PuO ₂	1.13	99.78	0.02	1.00	99.78	0.02	1.00	99.78	0.02
K	1572.80	K ₂ O	1.21	1895.22	0.34	0.89	1695.47	0.35	1.00	1695.47	0.38
Re	0.00	Re ₂ O ₃	1.13	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Rh	0.00	Rh ₂ O ₃	1.23	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Ru	0.29	Ru ₂ O ₃	1.23	0.35	0.00	1.00	0.35	0.00	1.00	0.35	0.00
Sb	0.00	Sb ₂ O ₃	1.20	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Se	0.00	SeO ₂	1.41	0.00	0.00	1.00	0.00	0.00	1.00	0.00	0.00
Si	75717.80	SiO ₂	2.14	161960.37	29.23	1.00	161960.37	33.36	1.00	161960.37	36.66

Table 3-4. Tank 241-C-106 Waste Composition. (2 sheets)

Element	Element wt kg	Oxide factor	Oxide wt kg	Oxide wt %	Water wash factor	Water wash solids, kg	Water wash wt %	Causic wash factor	Causic wash solids, kg	Causic wash wt %
Ag	563.40	Ag ₂ O	1.07	605.09	0.11	1.00	605.09	0.12	1.00	605.09
Na	124871.20	Na ₂ O	1.35	168326.38	30.38	0.61	102948.41	21.20	1.00	102948.41
Sr	109.80	SrO	1.18	129.89	0.02	1.00	129.89	0.03	1.00	129.89
SO ₄	2456.40	SO ₄	1.00	2456.40	0.44	0.44	1080.82	-0.22	1.00	-1080.82
Te	0.00	TeO ₂	1.25	0.00	0.00	1.00	0.00	0.00	1.00	0.00
Th	0.00	ThO ₂	1.14	0.00	0.00	1.00	0.00	0.00	1.00	0.00
Ti	0.00	TiO ₂	1.67	0.00	0.00	1.00	0.00	0.00	1.00	0.00
Ti	0.00	Ti ₂ O ₃	1.12	0.00	0.00	1.00	0.00	0.00	1.00	0.00
U	436.00	U ₃ O ₈	1.18	514.04	0.09	0.98	502.58	0.10	1.00	502.58
V	0.00	V ₂ O ₅	1.79	0.00	0.00	1.00	0.00	0.00	1.00	0.00
Zn	49.40	ZnO	1.24	61.45	0.01	1.00	61.45	0.01	1.00	61.45
Zr	2310.30	ZrO ₂	1.35	3118.91	0.56	0.95	2956.41	0.61	1.00	2956.41
Total Cation	343823.60	Total Ox.		554115.62	100.00		485519.90	100.00		441825.54
Water	559792.00									
Undiss. Solids (fusion)	27887.00									
Total Anions (CO ₃)	58961.30									
Total Sludge, kg	990463.90									
Tank Est., kg	1066272.00									
Mass Bal. %	92.89									
Missing:	Ox., Hydrex									

3.5 NOMINAL BLEND COMPOSITION

A nominal feed composition was developed for glass formulation purposes. This blend consists of 80 percent of the wastes in Tanks 241-AZ-101 and 241-AZ-102, 37 percent of the waste in Tank 241-C-106, and 15 percent of the waste in Tank 241-AY-102. The applicable sludge washing factors were estimated from the mass weighted average for each analyte in the nominal waste. Waste inventory estimates for this blend are presented in Table 3-5. This table also provides the estimated concentration of each component based on a concentration of 31 g/L washed oxide in the nominal feed. Since the water washed feed will be stored for an indeterminate amount of time, a small amount of sodium nitrite must be added to the waste for corrosion protection of the tanks. The amount of nitrite added (as NaNO_2) is approximately equal to 3 g/100 g of waste oxide or 0.93 g/L NO_2 to the water washed sludge.

Table 3-5. Nominal Blend Composition.
(0.8*AZ-101+0.8*AZ-102+0.37*C-106(Weiss)+0.15*AY-102)

Element	Element wt kg	Oxide	Oxide factor	Oxide wt kg	Oxide wt%	Water wash factor	Water wash solids, kg	Water wash solid, wt%	Water wash solids concentration, g/L	Caustic wash factor	Caustic wash solids, kg	Caustic wash solids, wt%	Caustic wash solids concentration, g/L	Water wash conc., g/L (element basis)
Al	39651.80	Al ₂ O ₃	1.89	74902.25	16.95	1.00	74902.25	18.90	5.86	0.52	38949.17	10.90	3.38	3.10
As	245.30	As ₂ O ₃	1.32	323.80	0.07	1.00	323.80	0.08	0.03	1.00	323.80	0.09	0.03	0.02
Ba	2217.60	BaO	1.12	2474.84	0.56	1.00	2474.84	0.62	0.19	1.00	2474.84	0.69	0.21	0.17
Be	6.46	BeO	2.78	17.93	0.00	1.00	17.93	0.00	0.00	1.00	17.93	0.01	0.00	0.00
Bi	197.50	Bi ₂ O ₃	1.12	221.20	0.05	0.97	214.56	0.05	0.02	1.00	214.56	0.06	0.02	0.01
B	210.40	B ₂ O ₃	3.21	674.75	0.15	1.00	674.75	0.17	0.05	1.00	674.75	0.19	0.06	0.02
Ca	5994.10	CaO	1.40	8385.75	1.90	1.00	8385.75	2.12	0.66	1.00	8385.75	2.35	0.73	0.47
Cd	4689.40	CdO	1.14	5355.29	1.21	1.00	5355.29	1.35	0.42	1.00	5355.29	1.50	0.46	0.37
Ce	428.20	CeO ₂	1.23	525.83	0.12	1.00	525.83	0.13	0.04	1.00	525.83	0.15	0.05	0.03
Cl	151.87	Cl-	1.00	151.87	0.03	0.64	97.20	0.02	0.01	1.00	97.20	0.03	0.01	0.01
Cr	1189.50	Cr ₂ O ₃	1.46	1737.86	0.39	0.87	1506.72	0.38	0.12	0.89	1340.98	0.38	0.12	0.08
Co	564.20	Co ₂ O ₃	1.41	793.83	0.18	1.00	793.83	0.20	0.06	1.00	793.83	0.22	0.07	0.04
Cu	214.80	CuO	1.25	268.93	0.06	1.00	268.93	0.07	0.02	1.00	268.93	0.08	0.02	0.02
Dy	27.90	Dy ₂ O ₃	1.15	32.00	0.01	1.00	32.00	0.01	0.00	1.00	32.00	0.01	0.00	0.00
F	924.10	F-	1.00	924.10	0.21	0.62	572.94	0.14	0.04	1.00	572.94	0.16	0.05	0.04
Fe	70937.50	Fe ₂ O ₃	1.43	101440.63	22.96	1.00	101440.63	25.59	7.94	1.00	101440.63	28.38	8.80	5.55
La	1810.50	La ₂ O ₃	1.17	2123.72	0.48	1.00	2123.72	0.54	0.17	1.00	2123.72	0.59	0.18	0.14
Pb	1304.20	PbO	1.08	1404.62	0.32	1.00	1404.62	0.35	0.11	1.00	1404.62	0.39	0.12	0.10

WHC-SD-WM-TI-768
Revision 0

Table 3-5. Nominal Blend Composition.
(0.8*AZ-101+0.8*AZ-102+0.37*C-106(Weiss)+0.15*AY-102)

Element	Element wt kg	Oxide	Oxide factor	Oxide wt kg	Oxide wt%	Water wash factor	Water wash solids, kg	Water wash solid, wt%	Water wash solids concentration, g/L	Caustic wash factor	Caustic wash solids, kg	Caustic wash solids, wt%	Caustic wash solids concentration, g/L	Water wash conc., g/L (element basis)
Li	20.80	Li2O	2.15	44.76	0.01	1.00	44.76	0.01	0.00	1.00	44.76	0.01	0.00	0.00
Mg	3003.20	MgO	1.66	4979.31	1.13	1.00	4979.31	1.26	0.39	1.00	4979.31	1.39	0.43	0.23
Mn	2073.90	MnO2	1.58	3280.91	0.74	1.00	3280.91	0.83	0.26	1.00	3280.91	0.92	0.28	0.16
Mo	33.80	MoO3	1.50	50.70	0.01	1.00	50.70	0.01	0.00	1.00	50.70	0.01	0.00	0.00
Nd	1399.40	Nd2O3	1.17	1633.10	0.37	1.00	1633.10	0.41	0.13	1.00	1633.10	0.46	0.14	0.11
Ni	3317.70	NiO	1.27	4223.43	0.96	1.00	4223.43	1.07	0.33	1.00	4223.43	1.18	0.37	0.26
P	1981.80	P2O5	2.29	4540.30	1.03	0.92	4181.62	1.06	0.33	0.34	1421.75	0.40	0.12	0.14
Pu	52.70	PuO2	1.13	59.60	0.01	1.00	59.60	0.02	0.00	1.00	59.60	0.02	0.01	0.00
K	2217.40	K2O	1.21	2671.97	0.60	1.00	2671.97	0.67	0.21	1.00	2671.97	0.75	0.23	0.17
Re	27.80	Re2O3	1.13	31.39	0.01	1.00	31.39	0.01	0.00	1.00	31.39	0.01	0.00	0.00
Rh	181.20	Rh2O3	1.23	223.42	0.05	1.00	223.42	0.06	0.02	1.00	223.42	0.06	0.02	0.01
Ru	568.10	Ru2O3	1.23	698.76	0.16	1.00	698.76	0.18	0.05	1.00	698.76	0.20	0.06	0.04
Sb	525.10	Sb2O3	1.20	628.54	0.14	1.00	628.54	0.16	0.05	1.00	628.54	0.18	0.05	0.04
Se	583.50	SeO2	1.41	819.82	0.19	1.00	819.82	0.21	0.06	1.00	819.82	0.23	0.07	0.05
Si	30315.0	SiO2	2.14	64845.28	14.68	1.00	64845.28	16.36	5.07	1.00	64845.28	18.14	5.63	2.37
Ag	526.0	Ag2O	1.07	565.57	0.13	1.00	565.57	0.14	0.04	1.00	565.57	0.16	0.05	0.04
Na	89457.30	Na2O	1.35	120588.44	27.29	0.66	79347.19	20.02	6.21	1.00	79347.19	22.20	6.88	4.61
Sr	213.60	SrO	1.18	252.69	0.06	1.00	252.69	0.06	0.02	1.00	252.69	0.07	0.02	0.02
SO4	6213.90	SO4	1.00	6213.90	1.41	0.47	2939.17	0.74	0.23	1.00	2939.17	0.82	0.25	0.23
Te	457.00	TeO2	1.25	572.62	0.13	1.00	572.62	0.14	0.04	1.00	572.62	0.16	0.05	0.04
Th	239.60	ThO2	1.14	272.66	0.06	1.00	272.66	0.07	0.02	1.00	272.66	0.08	0.02	0.02
Ti	148.80	TiO2	1.67	248.20	0.06	1.00	248.20	0.06	0.02	1.00	248.20	0.07	0.02	0.01
Ti	1308.80	Ti2O3	1.12	1461.93	0.33	1.00	1461.93	0.37	0.11	1.00	1461.93	0.41	0.13	0.10
U	4296.30	U3O8	1.18	5065.34	1.15	1.00	5065.34	1.28	0.40	1.00	5065.34	1.42	0.44	0.34
V	12.50	V2O5	1.79	22.31	0.01	1.00	22.31	0.01	0.00	1.00	22.31	0.01	0.00	0.00
Zn	134.00	ZnO	1.24	166.70	0.04	1.00	166.70	0.04	0.01	1.00	166.70	0.05	0.01	0.01
Zr	11811.30	ZrO2	1.35	15945.26	3.61	1.00	15945.26	4.02	1.25	1.00	15945.26	4.46	1.38	0.92
		Total		441866.10	100.00		396347.84	100.00	31.01		357469.15	100.00	31.01	31.01

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4.0 VITRIFICATION

Vitrification is a candidate process for converting the Hanford Site's HLW into a stable and durable (nonleachable) glass that is acceptable for disposal in a geological repository. The DOE has determined that the HLW disposal mission will be completed in two phases: (1) a Phase I demonstration where 3 to 6 percent of the HLW will be processed over a period of 5 to 9 years, and (2) a Phase II program where the remaining inventory of wastes will be treated in commercial scale facilities. The DOE has requested proposals for development of privately owned process facilities for Phase I HLW immobilization. The Phase I RFP provides for completion of design (December 31, 1999), starting hot operations of the HLW immobilization facility (June 1, 2007), processing the minimum quantity of HLW (June 1, 2011), and decontamination and decommissioning and RCRA closure (June 1, 2013).

4.1 IMPACT OF VITRIFICATION TECHNOLOGY ON FEED PROCESSABILITY

The design and operating characteristics of the glass melter are expected to have a significant effect on processability of Hanford Site waste. Important design considerations include the normal operating temperature of the melter, materials of construction, energy source, use of mechanical or natural convection, mixing, cold cap coverage, potential volatility effects imparted to the melt, characteristics of the feed system, and type of glass pour system employed. These features will generally have a pronounced effect on the operating characteristics of the melter, properties of the glass, and glass-production rates obtained. Certain glass properties such as viscosity, electrical conductivity, and density are directly affected by composition and temperature of the molten glass. Other properties such as WOL and solubility of glass property-limiting components in the melt are also affected by temperature and composition of the glass. The volatility of ruthenium and cesium oxides, mercury, and sulfur and cadmium oxides are affected by temperature of the glass, cold cap coverage, REDOX control, and agitation. All of these aspects must be taken into consideration when formulating the optimum glass composition and assessing the processability of Hanford Site waste.

4.2 STATUS OF VITRIFICATION TECHNOLOGY

DOE has decided that the HLW immobilization at the Hanford Site will be performed in privately owned and operated facilities. The purpose of this study is to summarize and evaluate existing waste vitrification data and glass industry and DOE vitrification experience with similar wastes. The evaluation in this section focuses on melter feed processability, product/process control, state of technology development, facility/system integration, design life of the melter, melter system scaleup, product sampling and recycle, secondary waste streams, operability, factors affecting total cost, and features creating special or unusual safety or environmental problems.

4.3 HIGH-LEVEL WASTE MELTERS

Several different melters have been extensively investigated for the Hanford Site HLW program. This includes both low-temperature (1,150 °C) and high-temperature (1,350 °C) joule-heated melters. A considerable number of melter tests have been performed with simulants and a few tests with samples of radioactive tank waste. Laboratory-scale tests with radioactive waste are needed to establish a link between simulant based studies and the behavior of actual tank waste sludges. Use of simulants needs to be verified by comparing the behavior of simulants to actual tank waste during feed preparation and vitrification of HLW. This comparison not only helps to validate testing results with simulants, but also provides a basis for validating glass property models used to formulate acceptable glass compositions. In the following section, the most important features and typical range of operating conditions will be described for each melter, together with a summary of results obtained from tests with radioactive wastes.

4.3.1 Low-Temperature, Joule-Heated Melter (LTM)

Joule-heated melters are typically heated by passing alternating electrical current through glass using submerged electrodes. The alternating current is conducted between two or more electrodes by mobile alkali metal ions (sodium or lithium) in the molten glass. Joule-heated melter designs vary with most employing a two-electrode arrangement. Others have a third electrode at the bottom to control the temperature of the bottom glass. The melter surface geometry is normally specified by the required surface area to meet production requirements and the depth required for adequate residence time. Glass residence times are typically determined by component volatility concerns (sodium metaborate volatility) and canister glass homogeneity considerations, that favor longer residence times for the HLW glass.

4.3.1.1 Melter Design. Joule-heated melters have certain advantages in comparison to other melters, especially the ability to suppress the volatilization of melt constituents and to melt opaque or "black" glasses where radiant heat transfer for indirect heating is limited. This melter is normally limited to a temperature range determined by the materials of construction for the refractories and electrodes. For the low-temperature melter, this temperature range is usually limited to about 1,150 °C because of temperature limits imposed on the system by the submerged Inconel¹ 690 (high-chromium-nickel alloy) electrodes.

Most joule-heated melters are ceramic lined with Monofrax K-3 fused cast refractory or equivalent (chrome alumina-zirconia-silica), although a few designs are Inconel 600 or 690 metal lined. The metal melters either operate at a lower temperature (<1,200 °C) or are water cooled such that the metal liner is protected by a layer of frozen glass. Because glass is not electrically conductive at room temperature, joule-heated melters must have auxiliary

¹Inconel is a trademark of Inco Alloys International, Incorporated.

heating for initial startup. This is usually provided by electrical resistance heaters placed in the plenum space above the glass surface.

It is assumed that the HLW melters will most likely be slurry fed in a fashion similar to the Savannah River Defense Waste Processing Facility (DWPF) melter and designed HWVP melter. Low-temperature, joule-heated melters have been typically designed to accept a slurry feed (although calciner- and dry-fed melters have been evaluated as well). Depending on the size of melter, more than one slurry feed nozzle may be required to ensure adequate coverage of the melt surface. Significant work has been done to develop feed nozzles that will not clog. Current feed nozzles are water cooled to avoid plugging. The nozzles can be water flushed whenever the feed is shut off to ensure that the feed does not dry in the nozzle.

There are several variations of this melter design (sloped bottom, dry fed with an evaporator, agitated, etc). The flat-bottom melter is normally considered to be the reference case because most testing in this country has been conducted in flat-bottom melters. It is assumed that a low-temperature melter (LTM) will be equipped with a bottom drain and a differential pressure (vacuum) overflow system for discharging the glass. The side-entering overflow drain can be machined from a single block of Monofrax E or K-3 refractory and heated by passing an alternating current through the Monofrax E refractory or by using an array of silicon carbide heaters in the pour section of the overflow drain. The bottom drain consists of an inductively heated vertical pipe that is enclosed in a hollow copper-tube coil that serves as an induction heater and cooling coil. The drain operates as a freeze valve. When glass is to be poured, the drain pipe is inductively heated to melt the glass plug. The glass-pouring rate is essentially controlled by varying the induction power. When enough glass has been poured, the induction power is reduced until glass flow stops. The ability to adequately stop glass flow is determined by certain characteristics of the system, including the level of glass (or differential pressure at the bottom of the melter), the density and viscosity of the glass, and the diameter of the pour stream or throat diameter of the pour spout. These characteristics, in addition to the basic materials used, are important to the safe and reliable operation of an inductively heated bottom drain.

The ceramic-lined, joule-heated melter has been selected for HLW vitrification in the United States (DWPF, West Valley Demonstration Project [WVDP]), Belgium, Germany, Japan, and Russia. These melters have two or more layers of refractory inside a water-cooled containment box. The glass-contacting refractory is usually high-chromia, fusion-cast refractory (K-3). The corrosion rate for K-3 is 10 to 20 $\mu\text{m}/\text{day}$ in the most aggressive areas at the melt line, but more typically 4 to 8 $\mu\text{m}/\text{day}$ as experienced by Pacific Northwest National Laboratory (PNNL) in the pilot-scale ceramic melter (PSCM). The fused-cast refractory in the HWVP melter design is 30-cm thick. Assuming a maximum corrosion rate of 20 $\mu\text{m}/\text{day}$, the expected refractory life would be 42 years. The electrical conductivity of glass-contacting refractory is important. If the refractory is too conductive (chromium content is too high), short circuits may develop between the electrodes via the refractory. This is only a concern with low-conductivity (low-alkali content) glasses. Because the

fused-cast refractory is not very insulative, it is usually backed with layers of more insulative ceramics such as high-alumina-castable ceramics or ceramic board.

High-chromium alloys are typically used for bubblers and thermocouple wells to monitor density and temperature of the glass. These alloys are susceptible to chloride and sulfate molten salt attack, with chloride dissolving the protective oxide film and sulfate induced sulfidation of the base metal (Marra 1994). Inconel 690 catastrophically failed under these conditions in the PAMELA melter (Bickford 1986). Sulfidation of these alloys can only be prevented by avoiding the accumulation of molten salts in the glass and cooling these alloys below a critical temperature of 650 °C for nickel based alloys.

A failure modes and effects analysis (FMEA) was conducted for the West Valley vitrification system in 1987. This analysis identified nineteen items that had a high or medium failure frequency with lengthy repair times. Most of the items identified were either external to the melter system (feed pump) or were melter components that are easily repaired or replaced (feed nozzle, thermowells). A few were identified whose failure could require replacement of the melter or significant repair periods. These include failure of the bottom electrode due to noble metals accumulation and glass discharge pluggage. The DWPF and HWVP designs are based on remote canyon facilities where access to equipment occurs from the top and all maintenance is accomplished with cranes and impact wrenches and not with manipulators. This leads to melter designs that have all access from the top because horizontal movement is difficult. The typical facility design calls for a valve corridor as a supplement to the canyon facility.

4.3.1.2 Glass Formulation. For LTM, the waste form is assumed to be borosilicate glass. Borosilicate glass has been recommended based on its ability to immobilize a wide range of wastes, its insensitivity to radiation damage and general properties such as thermal stability, chemical durability, and processability. The acceptable range of glass composition is normally limited to the range of compositions previously evaluated for the HWVP Project (the composition variability study [CVS] range of compositions for which empirical glass property models are considered valid). However, some compositions in the CVS range may fail to meet all of the glass property constraints while others outside the CVS range could satisfy these requirements. Most CVS composition information was developed for NCAW, which is very similar in composition to the proposed Privatization Phase I demonstration feed.

Iron, chrome, and nickel spinels are usually the first crystals to form in the NCAW glass upon cooling. Sometimes, especially with high-zirconium feeds, zirconia-containing crystals also may form in the melt or in annealed samples of glass. The repository waste acceptance preliminary specifications (WAPS) do not limit the amount of crystalline material in the glass. However, spinel phases and other crystalline phases that may precipitate in the glass are of concern because they may reduce the durability of the glass or affect the operation of the melter by filling the melt cavity or pour spout with a crystalline phase sludge. Glass compositions for high-waste-oxide-loaded glasses typically are limited by the concentration of crystal-forming components in the glass. HLW glasses are typically tested

by annealing the sample for 24 hours at 100 °C below the nominal operating temperature of the melter (viscosity of 4 Pa-s). Glasses are generally considered to be acceptable if they have less than 0.1 vol% crystals after annealing.

In general, the following glass properties are also considered to be necessary for acceptable HLW glass:

- REDOX potential (determined by ferrous to ferric [$\text{Fe}^{+2}/\text{Fe}^{+3}$] iron ratio) needed to control oxygen release and foaming tendency in the glass and for overly reduced glasses, precipitation of noble metals, spinel phases, and sulfide species. The $\text{Fe}^{+2}/\text{Fe}^{+3}$ ratio should range from 0.005 to 0.3 for acceptable REDOX control.
- Viscosity controlled in the range of 2 to 10 Pa-s (20 to 100 poise) to limit refractory erosion-corrosion on the low side and to ensure adequate mixing and reasonable melt rates below the upper limit.
- Maximum liquidus temperature (temperature at which crystalline phases appear) set 100 °C below the nominal operating temperature of the melter so that crystals will not form in the melt.
- Electrical conductivity controlled to allow joule heating of the glass. This aspect depends on the distance between electrodes and acceptable current densities and localized corrosion rates of the electrodes. Electrical conductivity at 1,150 °C should be in the range of 0.18 to 0.5 S/cm, while the current density of the Inconel electrodes should be significantly below 1.6 A/cm² and more typically near 0.8 A/cm².
- Durability as determined by the acceptable leaching rate of the most soluble component in the glass (sodium and boron) compared to the EA glass, which is the reference durability standard for HLW glass (WAPS.DOE/EM-0093.Rev.1. May 1995).

Waste loading is defined as the weight percent of waste (oxides) in the HLW glass. This value is extremely important because it determines the required production rate of the melter and the total amount of glass produced from Hanford Site waste. Glass quantities are important for several reasons. First, the repository charges are likely to be in the range of \$357,000 for the standard 0.6-m diameter by 3.05-m tall canister, based on the present cost allocation between civilian and defense wastes (DOE 1995). This allocation assumes 9,860 canisters of HLW glass from the Hanford Site, although 13,800 standard canisters (at 51 percent WOL) could be produced and larger 0.6-m diameter by 4.5-m tall canisters may be used (Orme 1995). Overall, nearly \$5 billion in cost savings could be realized by increasing the WOL of Hanford Site waste from 25 to 51 wt% based on the standard size canister. Second, processing of each canister in the vitrification plant is very expensive.

Each canister must be cooled, welded close, decontaminated, and stored. This operation is likely to be one of the most costly operations in the plant.

4.3.1.3 Melter Performance Sensitivities. Compounds or elements that may require special treatment in the offgas include mercury, carbon-14 (^{14}C), and iodine-129 (^{129}I). These elements will not be incorporated into glass at 1,150 °C, except for a minor amount of mercury, and must be captured in the offgas system and sent out as a secondary waste stream. Carbon-14 probably will be released to the atmosphere as long as *Clean Air Act Amendments of 1990* standards are met. Chloride, fluoride, technetium, cesium, cadmium, and ruthenium oxide are partially soluble in glass but require special consideration in the offgas treatment system for recycle. These components are routinely scrubbed out of the melter offgas and may be recycled to the melter feed.

Noble metal oxides and alloys of rhodium, palladium, silver, and ruthenium are virtually insoluble in borosilicate glass. Previous tests and studies have indicated that accumulation of these metal oxides can cause premature failure of the melter due to electrical short circuiting even at concentrations as low as 0.1 wt% in glass. Designs developed by the Germans (Kernforschungszentrum Karlsruhe GmbH [KfK]) have proven that accumulation of noble metals in the bottom of the melter can be avoided by having a steeply sloped (75 degrees) bottom that allows the precipitated noble metal particles to be drained out of a bottom drain.

The Japanese have developed a method that limits the deposition of noble metals by cooling the glass near the melter floor (Elliott 1994). The cold, viscous glass near the bottom acts as a barrier and significantly slows the settling rate of noble metal particles. Therefore, noble metals remain in the molten glass above the layer of cold glass that is maintained at a temperature of 850 to 950 °C. Such glasses can be poured through the bottom drain of the melter when the bottom temperature is raised from 900 to 1,050 °C to reduce the viscosity of the glass. With this method, the Japanese have demonstrated that ruthenium oxide accumulation rates could be reduced from 43 to 17 percent and PdO from 39 to 3 percent, with some tests indicating less than 1 percent of the noble metals having accumulated in the melter.

The Phase I water washed nominal blend contains 0.38 wt% noble metals (Ag_2O , Ru_2O_3 , Rh_2O_3 , TeO_2 , PdO). At 48 percent WOL in the glass, the noble metals concentration should be 0.182 wt%. If the Phase I feeds are processed separately, Tank 241-AZ-101, 241-AZ-102, and 241-C-106 glasses should nominally contain 0.17, 0.11, and 0.24 wt% noble metal oxides, respectively. Recent grab samples from Tank 241-C-106 indicate the bounding noble metal oxide concentration could be as high as 0.52 wt% in glass because of the amount of silver and ruthenium in this waste. This noble metals concentration is likely to cause premature failure of the melter electrodes and bottom refractory (based on the agglomeration and alloying characteristics of low melting temperature of silver (960 °C) and laboratory-scale tests performed with feeds containing from 0.33 to 0.46 wt% noble metals). An added concern is the high silver inventory in Tank 241-AY-102, with 1.85 wt% silver oxide (Ag_2O). Glass samples from the research scale melter (RSM) test performed in 1991

by PNNL contained from 0.332 to 0.465 wt% noble metal oxides, with accumulation rates of 5 to 46 percent of the incoming feed, depending on the concentration of noble metals. The electrodes and bottom refractory of the RSM were badly damaged, in all likelihood by accumulation of noble metals from the 0.465 wt% noble metals feed. RSM glass samples taken from the bottom of RSM contained clusters of RuO_2 needles and metallic particles of Ru/Rh surrounded by layers of Pd/Ag/Te alloy and RuO_2 (Cooper et al. 1994).

Semi-volatile elements are only partially soluble in glass. Table 4-1 shows typical melter decontamination factor (DF) for elements with DFs below 300, except HgO . Two different values are shown for HgO to illustrate the effect of glass composition on melter DFs for highly volatile components.

The DF is defined as the ratio of the mass flow of an element into the melter divided by the mass flow in the offgas. The DFs shown in Table 4-1 were determined for liquid-fed melters operating at 1,150 °C. Volatility losses from liquid-fed melters are affected by certain operating parameters (e.g., cold cap coverage, plenum temperature, and feed/glass composition). Because of its high-specific activity, cesium volatility is of particular interest. Cesium is believed to volatilize as cesium oxide, cesium hydroxide, or other compounds (Erlebach 1960, IAEA 1982). If chloride is present, cesium chloride is assumed to be the primary volatile species (Burkholder and Allen 1987). The presence of technetium has been shown to increase cesium volatility, as well, due to formation of cesium technetium oxide. Upon addition of technetium to a feed, cesium losses increased from 2-5 to 18 percent (Baumgartner 1984).

Radioactive materials are confined to the melter and offgas system because the melter plenum and overflow are kept under slight vacuum to prevent the airborne release of such materials to the melter cell. However, deposits can still accumulate in the offgas line. This can be controlled by using a film cooler to quench and dilute the hot particulate material from the melter or by designing the offgas line so that deposits can be physically removed with a reamer, as was done for the PAMELA melter.

Joule-heated melters typically have side- or bottom-entering electrodes constructed of Inconel 690. Submerged Inconel-690 electrodes corroded at an average rate of 3 μ /day in the PSCM after 3 years of operation (Barnes and Larson 1981). Inconel materials in the plenum space have been severely attacked due to halides and sulfates in the feed stream (Dierks 1980). Corrosion attack by sulfur and its compounds has been found to be a potentially severe problem. In the PAMELA plant, the baffle plate and the wall opposite the offgas inlet corroded through in less than 150 hours (Chapman 1983). The material used in this area of the PAMELA was the European equivalent of 304 stainless steel (SS). Inconel 690 does provide some improvement in time to failure, but the performance characteristics of this alloy have not been determined over the projected life of the plant. Most Inconel materials exposed to offgas (thermowells and bubblers) can be replaced remotely.

Table 4-1. Liquid-Fed Ceramic Melter Decontamination Factors.

Oxide	Wt% oxide in waste	Melter DF	Reference ^a
CdO	1.03	250	Perez 1994
Cl	0.16	4	Perez 1994
Cs ₂ O	0.15	100-200	Perez 1994
F	0.03	4	Perez 1994
HgO	0.0005	2833.3	Pegg et al. 1994
HgO	0.008	1.3	Chapman 1987
Rb ₂ O ₃	0.007	290.0	Perez 1994
Rh ₂ O ₃	0.013	29.0	Whittington 1993
Ru ₂ O ₃	0.047	10.0	Scott 1985
SO ₃	0.341	6.4	Goles and Nakaoka 1990
SeO ₂	0.01	4	Perez 1994
Tc ₂ O ₇	0.021	2.5	Chapman 1987
TeO ₂	0.003	11.0	Scott 1985

DF = Decontamination factor

^aSee Section 8.0, "References," for complete reference citations.

If the melter plenum is maintained at high temperature, it may be assumed that organics are mostly destroyed in the plenum space of the melter. The plenum temperature is normally about 650 °C when supplemental plenum heaters are used. The destruction and removal efficiency (DRE) for formic acid was recently determined for liquid-fed ceramic melters. The destruction of formate was at least 99.8 percent for feeds containing 37 to 46 g/L formate (Perez 1994). Similar tests have been performed at SRS to assess destruction of organics in the DWPF melter (Crow 1984, 1987).

Joule-heated melters have been demonstrated with nonradioactive simulants at SRS (scale-glass melter with 1.2-m² surface area and melt rates of 45 kg/h), at West Valley Nuclear Services (melter of 2.15-m² area and melt rates of 45 kg/h), and at PNNL (LFCM of 1.05-m² surface area and melt rates of 25 to 30 kg/h, and the PSCM of 0.76-m² surface area and melt rates of 22 kg/h).

Existing test facilities, in addition to those previously mentioned, are the K-6 melter in Germany (0.88 m²), the advanced B melter in Japan (0.88 m²), the Integrated DWPF Melter System (IDMS) melter at SRS (0.29 m²), the 1/10th-scale (0.25 m²) and PSHTM (2 m²)

melts at PNNL, and finally the two Duramelts (0.13 and 0.33 m²) at GTS Duratek. Remote operation of LTM's in radioactive environments have been demonstrated in Russia (Mayak plant), Belgium (PAMELA plant), and the United States (radioactive liquid-fed ceramic melts at PNNL, SRS, and WVDP). Similar systems have been designed but not operated in Japan (Tokai) and the United States (HWVP). The SRS and WVDP production melts (2.5 m²) are currently being used to process radioactive wastes at their respective sites.

The reliability of a joule-heated melter can be expressed as the total online efficiency (TOE). This is defined as the total hours that a melter operates divided by the total hours that a melter is in place, including the time for melter changeout. A TOE of 60 percent was assumed for the HWVP design. This was arrived at by assuming the melter was online 70 percent of the time with a 6-month total shutdown every 3 years. This was considered to be a conservative design basis because efficiencies of greater than 85 percent have been regularly attained with test melts.

The average online efficiency of the PAMELA melter was 85 percent, while the efficiency was greater than 98 percent during the recent LFCM tests conducted by PNNL (Perez 1994). In practice, melter efficiencies of less than 70 percent should be expected during the first year of operation for a production plant with some increase in online efficiency as problems are worked out of the process. Because the waste compositions are likely to vary, some uncertainty needs to be assumed in the analysis and control of waste composition through the process and in the control of glass composition. Based on the experience at SRS, it is reasonable to assume that the HLW vitrification process will produce glasses of about 85 to 90 percent of the theoretical WOL. This offset is generally caused by the statistical uncertainties that must be taken into account in liquidus temperature predictions (both model and process uncertainties) (DWPF manual B-DD-S-00001).

4.3.1.4 Estimated Melter Design Life. The minimum melter lifetime of the HWVP melter was estimated to be 2 years and the expected lifetime 3 to 5 years. The IDMS melter at SRS has experienced very little refractory wear after 5 years of continuous idling (at 900 °C) and operation (at 1,150 °C). Monofrax E and K-3 refractories have shown little or no corrosion after 5 years of operation at WVDP. If refractory wear is the only consideration, ceramic-lined melts should last for 10 years or more under continuous service conditions (at 1,150 °C).

The most likely scenario for early melter failure is believed to be the accumulation of noble metals on the melter floor causing short circuiting of the lower set of electrodes. The potential accumulation of noble metals could be mitigated by one of several control strategies. One strategy is to blend the Phase I feeds to reduce the likely concentration of noble metals. Another approach developed at SRS involves adding water to the glass surface during idling periods to increase convective flow in the melter, thus limiting the settling of noble metals. A recent analysis of IDMS data from SRS suggests that 70 to 80 percent of the noble metals may have settled from the melt due to the absence of sufficient convection when the melter was idled (Hutson 1994).

Another important aspect relates to the composition and electrical characteristics of the glass found in the bottom of the IDMS melter. This glass apparently contained enough precipitated spinel that the glass was relatively insulating compared to other melter bottom glasses with similar concentrations of noble metals (PAMELA). In this case, the electrical characteristics of the bottom glass appear to be controlled by the composition of the melt. Still another strategy is to design a sloped-bottom melter and bottom-drain system for sludge-forming noble metal wastes or to devise a control scheme, such as the one developed by the Japanese, that allows for the efficient purging of noble metals. Agitation of the glass with mechanical mixers or air spraying may also reduce accumulation of sludge. These are important strategies to consider for noble metal wastes such as those in the Phase-I tanks.

4.3.1.5 Product/Process Control. The product quality, as such, will be determined by glass formulation. Glass development is done to ensure that the final product will be at least as durable as the reference EA glass when tested with the product consistency test (PCT) and Materials Characterization Center (MCC)-1 leach test. Waste product homogenization is ensured by maintaining a sufficient residence time in the melter. For the DWPF melter, the residence time averages 2 days. This long residence time may be necessary to ensure proper dissolution and mixing of the waste in the glass product (although adequate results could be obtained in a much shorter period of time for some wastes). Tracer tests have indicated the melter acts as a continuous stirred tank reactor so the waste should be well mixed with the glass-forming components in the melt. With a longer residence time, insoluble particles or crystals will have more opportunity to settle to the melter floor.

Certain measurements are required to ensure safe process control in a remote operating environment. This includes temperature measurements in the glass, plenum space, glass discharge areas, refractories, and cooling water circuits. The most difficult parameter to control remotely is the amount of cold cap coverage (i.e., fraction of melt surface covered with solid unmelted feed material). The feed must be controlled such that the melt pool is not completely flooded with feed causing potential bridging of the cold cap (i.e., adherence of the cold cap to the walls of the melter). Insufficient cold cap coverage will lead to increased volatilization and gas release, and excessive heat loss. The cold cap coverage is usually controlled based on past experience and plenum temperature (which increases as the cold cap is melted by hot glass) and visual inspection with a remote TV. Remote infrared cameras have been investigated by PNNL as a control option but have not been implemented to date. The melter plenum must be maintained at a slight vacuum to avoid contamination of the process cells. This control is dynamic because of the variable nature of the melter cold cap.

Joule-heated melters operate continuously under very steady conditions such that process upsets are rare. Glass production rates, however, can be adversely affected if foam develops in the glass melt. Gas bubbles are commonly formed when the REDOX state of the melt is abruptly changed with a rapid increase in electrode power and glass temperatures in the vicinity of the electrodes. This effect can be suppressed by adding chemical reductants such as formic and glycolic acid to the melter feed. During idle periods, water can be fed to the melter to cool the upper glass surface to minimize volatilization and to increase

convective mixing so noble metals will not settle in the melter. Ceramic-lined, joule-heated melters are based on a mature technology that has been tested since the early 1970's for the vitrification of HLW.

4.3.1.6 Data Needs. Most HLW glass formulations are being developed for high-waste-oxide-loaded glasses. For high-nickel, -chromium, or -iron feeds, spinel precipitation and sludge accumulation problems are likely to occur. Precipitation and settling characteristics of a spinel type phase need to be assessed under different operating conditions. Viscosity and electrical conductivity measurements are needed to determine whether spinel phase enriched sludge deposits are primarily a mechanical or electrical problem in the melter. Glass melts containing from 2 to 10 vol% crystals are known to have approximately Newtonian viscoelastic behavior (Plodinec 1986).

NCAW waste by definition is comprised of wastes from Tanks 241-AZ-101 and 241-AZ-102. Based on crucible tests to date, glasses with 33 wt% NCAW do not form spinel phase, while glasses with 40 to 50 wt% NCAW typically precipitate from 0.5 to 1.0 vol% spinel at 1,050 °C (100 °C below the operating temperature of the melter) (Fini and Hrma 1994, Hrma and Bailey 1994). At 50 wt% NCAW, baddeleyite (ZrO_2) appears in the glass and its concentration increases sharply with waste loading (Hrma and Bailey 1995). The WOL for NCAW glass could be increased by 30 to 40 percent if a practical sludge removal system is developed for the low temperature melter. Several different approaches may be potentially promising, including: (1) a cone-shaped melter bottom, bottom electrode and drain system, (2) a detachable side entering drain with features to induce flow, (3) a ceramic stir or bubbler system to mobilize the solids and the injection of low viscosity glass to purge the sludge, or (4) a top entering vacuum removal system, such as the one developed by PNNL for West Valley (evacuated canister design) (Barnes 1989). If sludge removal capabilities could be developed for the LTM, LTM glasses could be formulated to match, or possibly exceed the waste oxide loading capabilities of the HTM, significantly reducing the predicted life-cycle costs for the LTM.

4.3.2 High-Temperature, Joule-heated Melter (HTM)

The HTM operates on the same principal as other electric melters by passing an alternating electrical current through the glass to induce self heating into the molten glass. The HTM, as embodied in the most recent PNNL design, has two electrode systems, a pair of shielded and water-cooled Inconel/molybdenum electrodes inserted from the top, and a second set of Monofrax² E (high-chromium oxide) refractory electrodes incorporated into the melter's refractory envelope. Because these electrodes are capable of higher temperature performance, this melter can be typically operated at temperatures of 1,350 to 1,550 °C. It is anticipated that higher operating temperatures will lead to a corresponding increase in the melt rate and waste oxide loading potential for HLW glasses. The melt surface geometry and melter configuration are dictated by electrode spacing and surface area requirements, and by the glass residence time and production requirements of the melter (melt rate capacity).

4.3.2.1 Melter Design. The HTM refractory envelope designed by PNNL consists of Monofrax E and Monofrax K-3 fused cast refractory for the glass-contact zone in the melt cavity, surrounded by more insulative layers of high-alumina Alfrax-66 refractory and ceramic board (Duraboard-LD), and a water-cooled Inconel shell. The small-scale high temperature melter (SSHTM) test melter developed by PNNL uses a pair of Monofrax E electrodes spaced 38 cm apart. This melter is equipped with a differential pressure overflow system and an inductively heated bottom drain for glass pouring. The SSHTM overflow discharge uses electrical resistance heaters to maintain a molten pour stream. The bottom drain consists of an inductively heated, vertical pipe or udder inserted into a hole in the melter floor. This pipe is surrounded by a copper coil that serves as an induction heater and cooling coil. The drain valve assembly is typically sealed and connected to the glass canister with a metal bellows that can be filled with nitrogen to protect the drain valve materials.

The LTM and HTM designs are generally similar except for the electrodes. Candidate electrode materials are currently being evaluated for the HTM, including molybdenum, chromium, chromium-alumina, chromia, Monofrax E, platinum-10 percent rhodium, tantalum, tin oxide, tungsten, and graphite.

Molybdenum electrodes are typically used in most commercial electric melters except for those making glasses that contain species such as lead and arsenic that require more oxidizing conditions (Calmus 1995). Hanford Site HLW sludges contain lead and arsenic, as well as metals of lower oxidation potential, or ones that may form low-temperature melting alloys with molybdenum. Molybdenum electrode stems must be cooled or maintained in an inert atmosphere because molybdenum rapidly oxidizes in air above 600 °C. Several electrode holders have been designed to protect top entering molybdenum electrodes from oxidation in air. Tin oxide electrodes could be used in an oxidizing environment, but are limited to a maximum operating temperature of about 1,300 to 1,350 °C. These electrodes also have a lower maximum current density (0.5 A/cm² compared to 2 A/cm² for molybdenum), and typically require more elaborate bus bar connections because tin oxide is electrically insulating below 900 °C (Calmus 1995).

For oxides such as tin oxide, chromia and the chromia-alumina refractories (Monofrax E and K-3), simple dissolution (interdiffusion of components of the electrode and melt) is usually the dominate corrosion mechanism (Woolley 1995). Monofrax E electrodes are also susceptible to "rat-holing," where corrosion channels preferentially develop because of the large grain size and the nonuniform distribution of grain size and possible voids in the material. Recent PNNL tests have indicated that rat-holing is a severe problem with Monofrax E electrodes. Three of four electrodes apparently failed due to rat-holing in the SSHTM. Graphite is limited by its low allowable current density (0.2 A/cm²) and its incompatibility with oxidizing glasses. Platinum electrodes commonly degrade in a reducing environment or by forming low melting temperature alloys with impurities in the melt.

Refractory corrosion rates typically double with an increase of 50 to 100 °C in glass temperature (Woolley 1991). This general correlation only applies to glasses of similar composition and cannot be used to predict the corrosion behavior of higher viscosity glasses.

Corrosion rates are typically influenced by the temperature and viscosity of the glass (with lower viscosity glasses producing higher corrosion rates), current density of the electrodes, and the presence of reducible ions and molten salts in the melt. Corrosion results for candidate electrode materials are summarized in Table 4-2 (Freeman 1995). These results show that platinum-rhodium and Monofrax E are the preferred electrode materials for the HTM based on static corrosion tests.

Table 4-2. Corrosion Rates for Candidate Electrode Materials in
HTB-650 Glass at 1,400 °C.

Material	Applied current, A/cm ₂	Electrochem. corrosion, mm/year	Corrosion (7 day static), mm/year	Corrosion (24 h static) mm/year
Platinum-10% Rhodium	1.0	2.9		
Platinum-10% Rhodium	1.6		- 0.5	
Monofrax E	0.2	5.0	0.1	
Chromia	0.2	36.6	2.0	
Molybdenum	1.0	77.6	14.0	114.4
Tungsten	1.0	87.9	42.8	85.8
Tin Oxide	0.2	192.3	86.7	
Tantalum	1.0		> 133.0	332.8
Chromium	1.0		> 163.6	244.1
Chromium-alumina	1.0		> 151.7	> 1068

For the SSHTM melter, Monofrax E electrodes corroded at a rate of about 50 mm/year at the melt line and 25 mm/year below the melt line (general corrosion rate, exclusive of rat-holing effects, over an 8 month period at 1,350 °C). This rate was found to be much higher than the previous estimates from electrochemical and static tests because the electrodes were exposed to melter feed materials as they were being processed into glass. Monofrax K-3, however, corroded at a rate of about 9 mm/year in the SSHTM, or slightly less than the rate of 13.5 mm/year from static tests (Freeman 1995). Based on these rates, the useable lifetime for 30 cm of K-3 refractory should be about 20 to 25 years, but only several years for 30 cm of Monofrax E refractory in the HTM.

4.3.2.2 Glass Formulation. For the HTM, the waste form is assumed to be sodium-alumina-silicate high-temperature glass, where silica is the only glass former added, or borosilicate glass because of its acceptability as a recommended waste form. These glasses can be readily processed at 1,350 °C with acceptable viscosity, durability, and crystallinity (or liquidus temperature) properties in the melter. It is anticipated that HTM waste acceptance and glass property limits will be identical to the established limits for LTM glasses. HTM glass formulations should be based on identical constraints for REDOX, viscosity, electrical conductivity, liquidus temperature and durability.

Several glass formulations have already been developed for the HTM, including glasses based on NCAW and DST/SST blend, using CVS property and composition models (Kim 1994). These glasses were formulated to find the optimum frit composition (Na_2O , Li_2O , B_2O_3 , and SiO_2) for glasses with 50 wt% NCAW waste, and to find the maximum waste loading of DST/SST glass, in the range of 57 to 71 wt%, using SiO_2 as the only glass former.

For NCAW glass, it was determined that sodium and lithium oxides should be added in lieu of boric oxide to suppress the precipitation of spinel type phases and to increase the waste oxide loading of NCAW waste. The optimum glass contained 50 wt% NCAW, 44 wt% SiO_2 , and 6 wt% Na_2O , with a melting temperature of 1,350 °C and negligible crystallinity. DST/SST glasses were formulated to minimize precipitation of spinel phases and ZrO_2 . The optimum glass contained 62 wt% DST/SST blend waste and 38 wt% SiO_2 , with a melting temperature of 1,350 °C (at 4 Pa-s) and less than 0.5 vol% crystallinity (with 90 percent of the crystals being less than 1 micron in diameter). These glasses are of interest because they were used to formulate feeds for the SSHTM melter.

During FY 1994, two melter tests, SSHTM-1 and SSHTM-2, were completed using a simulated NCAW waste and blended DST/SST waste (Abrigo 1995b, Smith 1995). The SSHTM-1 test was divided into two segments with two separate feeds: (1) a 50 wt% NCAW feed using glycolic acid in lieu of formic acid as the chemical reductant and (2) a 50 wt% NCAW feed using only nitric acid (and no organic acid) to adjust the rheology of the feed. This test was performed to evaluate the use of glycolic acid as an alternate reductant (to suppress hydrogen production in the melter feed preparation process) and to assess the need for chemical reductants in the melter. The SSHTM-2 test consisted of four additional segments with two feeds: (1) a 47 wt% NCAW feed modified with boric and nitric acid and (2) a 65 wt% DST/SST feed simulant. This test sequence was performed to evaluate the performance of the HTM with a high-waste loaded glass. Additional tests, SSHTM-4 and SSHTM-5, were performed during FY 1995 to evaluate feed additives (boric acid), feed oxide loading, temperature effects and pouring characteristics of the new bottom drain (using a 1.9 cm instead of 1.27 cm orifice) (Lamar 1995).

4.3.2.3 Melter Performance Sensitivities. Certain elements and compounds cannot be incorporated into high-temperature glass. Such materials include ^{14}C , and ^{129}I . These must be captured in the offgas system and sent out as secondary waste. The exception is ^{14}C , that will probably be released to the atmosphere as long as *Clean Air Act Amendments of 1990* standards are met.

Noble metal oxides and alloys (rhodium, palladium, RuO_2 , and silver) are virtually insoluble in borosilicate glass. Previous studies have indicated that accumulation of these metals can cause premature failure of the low-temperature, ceramic-lined melter due to electrical short circuiting at concentrations as low as 0.1 wt% in glass (see Section 5.7.6).

During segment 1 of SSHTM-1, 1200 L of NCAW feed (with glycolic acid) were processed at temperatures of 1,350 to 1,400 °C, producing 520 kgs of glass (Abrigo 1995b, Smith 1995). Glass processing rates varied from 16 to about 26 kg/h-m². An insulating layer of scum formed on the surface of the glass toward the end of this test segment, but this layer was finally removed after plenum heaters were used to increase the surface temperature of the melt. Approximately 150 kgs of glass were produced during segment 2 of SSHTM-1, using nitric acid adjusted NCAW feed. The glass production rate averaged about 10 kg/h-m² at 1,350 °C. As indicated by the low production rate, nitric acid laden feed did not process well in the SSHTM. The feed formed a mound under the feed nozzle and did not spread out over the cold cap surface. Boric acid was added to the feed and within a few hours the feed was spreading over the cold cap and forming a flexible and highly desirable cold cap with active venting.

Boric acid modified NCAW feed (with about 3 wt% B_2O_3) was also used during the first segment of SSHTM-2, where the glass production rates increased to about 22 kg/h-m² because of enhanced fluidity of the cold cap. When the temperature was raised from 1,375 to 1,425 °C, the glass rate increased to about 30 kg/h-m². During segment 2 of SSHTM-2, approximately 1,700 L of blended DST/SST feed (HTB650) were introduced to the SSHTM and processed at 1,350 and 1,400 °C to assess the impact of temperature on processing rate and offgas composition. The DST/SST feed (with 0.3 wt% B_2O_3) tended to mound beneath the feed nozzle and formed a rigid cold cap at 1,350 °C, with an average glass production rate of about 22 kg/h-m². The glass production rate increased to about 28 kg/h-m² after the temperature was raised to 1,400 °C. During segment 3 of SSHTM-2, a ceramic agitator (resembling an Archimedes screw) was lowered into the melt and rotated at 40 to 60 rpm to enhance the processing rate of the SSHTM. It was determined that the glass production rate averaged about 20 kg/h-m² with agitation at 1,350 °C, or about the same rate as one might achieve without agitation. Melter performance results have not been fully analyzed and reported for test segments SSHTM-4 and SSHTM-5. Table 4-3 summarizes the glass production data currently available from the SSHTM.

Table 4-3. Glass Production Rates for the Small-Scale High-Temperature Melter.

Melter test	Feed type	Plenum heating	Melter surface area, m ²	Glass temp., °C	Glass production rate, kg/h-m ²
SSHTM-1,1	50% NCAW (gly.acid)390 g-TO/L	N	0.21	1,350	16
SSHTM-1,1	50% NCAW (gly.acid) 390 g-TO/L	N	0.21	1,400	26
SSHTM-1,2	50% NCAW (nit.acid) 380 g-TO/L	N/Y	0.21	1,350	10
SSHTM-1,2 SSHTM-2,1	50% NCAW (bor.acid) 341 g-TO/L	N	0.21	1,350	22
SSHTM-2,1	50% NCAW (bor.acid) 341 g-TO/L	N	0.21	1,375	21
SSHTM-2,1	50% NCAW (bor.acid) 431 g-TO/L	N	0.21	1,425	30
SSHTM-2,2	65% DST/SST (nit.acid) 535 g-TO/L	N	0.21	1,350	22
SSHTM-2,2	65% DST/SST (nit.acid) 535 g-TO/L	N	0.21	1,400	28
SSHTM-2,3	65% DST/SST (nit.acid) w agitator. 535 g-TO/L	N	0.21	1,350	20
SSHTM-4	Shim. NCAW All Blend (nit.acid)avg. 4a&4b	N	0.26	1,400	17.7
SSHTM-4a	Shim. NCAW All Blend ^a 409 g-TO/L	N	0.26	1,400	21.9
SSHTM-4b	Shim. NCAW All Blend ^a 272 g-TO/L	N	0.26	1,400	17.7
SSHTM-5	All Blend (Optima) ^b 376 g-TO/L	N	0.26	1,425	14.6

g-TO/L = grams of total oxide per liter.

^aShimmed NCAW: 9.3 g/l Al₂O₃, 8.21 g/l Fe₂O₃, 16.12 g/l Na₂O and 52 g/l SiO₂.

^bAll blend prepared by Optima Chemical: 31.8 g/l Al₂O₃, 26.4 g/l Fe₂O₃, 84.2 g/l Na₂O and 144.1 g/l SiO₂.

Table 4-4 provides comparable results for the LTM based on various PNNL melter campaigns (Abrigo 1995b).

Table 4-4. Glass Production Rates for Low Temperature Melters.

Melter test	Feed type	Plenum heating	Melter surface area, m ₂	Glass temp. °C	Glass production rate, kg/h-m ²
HBCM-86-2A	HWVP	N	0.25	1,150	29
HBCM-86-2B	HWVP	N	0.25	1,150	35
HBCM-86-2C	HWVP	N	0.25	1,150	29
LFCM-5	SRP	Y	1.05	1,150	18
LFCM-6	SRP	Y	1.05	1,150	14
LFCM-7B	SRP	Y	1.05	1,150	23
PSCM-3	SRP	N	0.73	1,150	22
PSCM-4	SRP	N	0.73	1,150	38
PSCM-15B	WV	N	0.73	1,135	27
PSCM-16	WV	N	0.73	1,160	19
PSCM-20	WV	N	0.73	1,175	18
PSCM-20	WV	N	0.73	1,175	14
PSCM-17	HWVP	N	0.73	1,150	9
PSCM-22A	HWVP	N	0.73	1,180	17
PSCM-22B	HWVP	N	0.73	1,180	19
PSCM-23	HWVP	N	0.76	1,150	28
LFCM-8A	HWVP	Y	1.05	1,170	28
LFCM-8B	HWVP	Y	1.05	1,160	26

HBCM = High-Bay Ceramic Melter

LFCM = Liquid-Fed Ceramic Melter

PSCM = Pilot-Scale Ceramic Melter

HWVP = Hanford Waste Vitrification Project (similar to Phase I feed)

SRP = Savannah River Plant (PUREX simulant similar to Phase I feed)

WV = West Valley Demonstration Project (commercial HLW feed simulant).

On average, the SSHTM produced about 22.9 kg/h-m² of glass during these trials (at 1,350 °C), while the LTM has typically produced about 20.8 kg/h-m² of glass at 1,150 °C. Based on these results, the melt rate enhancement features of the SSHTM (higher operating temperatures, more flexible glass formulations) did not produce the expected increase in glass throughput capacity anticipated for the HTM.

High waste oxide loaded glasses can also be formulated for the LTM by formulating glasses beyond the normal liquidus temperature limit for the LTM. For example, optimum formulations for the HTM might contain as much as 50 wt% NCAW waste (with 1.5 to 4.6 vol% crystals because of refractory corrosion, Table 4-7). Similar glasses can be formulated for the LTM that contain 33 wt% NCAW (with no evidence of crystallinity after 24 hours at 1,050 °C), and 40 to 45 wt% NCAW (with 0.4 to 0.9 vol% crystals after 24 hours at 1,050 °C) (Vienna et al. 1996, Fini and Hrma 1994). Dissolution of refractory components into glass increases aluminum and chrome levels and thus reduces the previously expected waste oxide loading advantage of the HTM (if corrosion source term components are included in calculating liquidus temperature limits for glass formulation). Refractory corrosion increased the predicted liquidus temperature of N508 glass from 1,162 to 1,346 °C during the SSHTM tests. This change was found to be consistent with the increase in crystallinity of the glass (from < 0.1 vol% crystallinity based on crucible tests to 4.59 vol% crystallinity in the melter product glass). For other wastes such as the all tank blend, the HTM appears to offer a 5 to 10 wt% oxide loading advantage over the LTM based on liquidus temperature properties of rare earth (cerium-neodymium) zirconates (Vienna and Hrma 1995).

During several of the recent SSHTM tests, glass production rates increased 30 to 35 percent when the temperature was raised from 1,350 °C to about 1,425 °C. The viscosity of the glass was also reduced from 7 Pa-s to about 3.7 Pa-s over this temperature range. This noticeable increase in the production capacity could be due to viscosity effects and the increase of convection in the melter, rather than to thermal effects in the cold-cap region. Similar affects were also observed during the RSM test where the melt rate increased (by a factor of 2.3) after the temperature was raised from 1,050 to 1,200 °C (and glass viscosity consequently reduced from 18 Pa-s to 4 Pa-s) (Research Scale Melter Test Report, comparing segments 4 and 7, [Cooper et al. 1994]). While viscosity manipulation techniques may be useful for enhancing the melt rate, these conditions could also lead to some acceleration of refractory corrosion in the melter. If the viscosity limits are eventually reduced, glass production rates could be increased in the LTM by reformulating the glasses for lower viscosity or by using electrodes capable of higher temperature performance in the LTM. The SSHTM tests also show that HTM feeds should had improved fluidity in the cold cap region with the addition of 3 to 5 wt% B₂O₃.

Semivolatile elements have a limited solubility in glass. Aerosol samples were taken during SSHTM-1 and SSHTM-2 to assess the impact of elevated temperature on the distribution of volatile and semivolatile components in the waste. Table 4-5 shows the melter decontamination factor (DF) for elements with DFs below 300. Volatility losses from liquid-fed melters are believed to be affected by cold cap coverage, plenum temperature, and the temperature and composition of the glass (presence of chlorides and other volatile species). From this data, it appears that volatility losses did not increase significantly when the temperature was raised from 1,150 to 1,400 °C.

Table 4-5. Melter Decontamination Factors for High-Temperature and Low-Temperature Melter (Abrigo 1995b).

Waste oxide	SSHTM-1,1 (HTM)	SSHTM-2,1/ SSHTM-2,2 (HTM)	LFCM-1 (LTM)	PSCM-23 (LTM)	LFCM-8 (LTM)
Ag ₂ O	80	28/365			60
CdO	195	52/2,029			250
Cr ₂ O ₃	200	21/491		440	317
Cs ₂ O		35/38	44	77	
K ₂ O	174	134/9,794			
Li ₂ O		53/136	940	1,500	
MoO ₃	120	11/599		110	196
P ₂ O ₅	156	415/2,487		2,600	54
SO ₃	43	6/334		6	10
ZnO	620	68/616		960	390
Cold Cap Coverage	90%	85%/95%			
Glass Temp. °C	1,400	1,390/1,350		1,150	1,160

HTM = High-temperature melter

LTM = Low-temperature melter.

Table 4-6 provides component volatility data from the low-level waste (LLW) melter tests conducted with Hanford Site's double-shell slurry feed (DSSF simulant) (Wilson 1996, Whyatt, Shade, and Stegen 1996). This information is of interest because it shows how volatility losses can vary with the design and operating characteristics of the melter. Babcock & Wilcox (B&W) used a slurry-fed cyclone combustion melter. The feed was atomized and sprayed into a cyclone combustion flame, with volatilization and high entrainment losses. Combustion furnaces of this type typically operate with an exit gas temperature of 1,800 to 3,000 °C. Westinghouse Science and Technology Center (WTSC) used a plasma torch-fired cupola furnace and atomized slurry feed, with significant volatility losses as expected due to high temperatures in the plasma jet. The U.S. Bureau of Mines (USBM) employed a carbon electrode melter using prereacted, dried, and pelletized feed as a batch blanket. Due to the high temperatures involved, volatility losses were high until steps were taken to increase the diameter of the electrodes and reduce the voltage to control volatility losses from the glass (by changing from arc furnace to joule-heating, thus reducing glass temperatures near the electrodes). Vectra demonstrated a high temperature (1,500 °C) joule-heated skull melter with fully calcined, dried and slurry feeds. Because of the angular placement of the electrodes, the central part of the melt pool was rarely covered with the cold cap, explaining the high volatility losses in Table 4-6. The glass melt rates with slurry feed were greater than rates with calcined or simulated calcine dry feeds. Duratek employed a low temperature (1,150 °C) joule-heated melter with Inconel electrodes, an air-lift overflow drain system and air bubblers to enhance convection and mixing in the melter. Average processing rates with slurry feed varied from 61 to 66 kg/h-m². (about three times the rate of the SSHTM with HLW feed). Volatility losses were moderately high because of the air bubblers and associated disruption of the cold cap. Envitco demonstrated a high temperature (1,450 to 1,500 °C) joule-heated melter with spray-dried and slurry feeds. Volatility losses were exceptionally low because of full cold-top batch coverage and the use of submerged, side-entering molybdenum electrodes that did not interfere with the cold cap.

Glasses produced during the SSHTM-1 and SSHTM-2 melter tests also contained much more crystalline material than anticipated from laboratory studies. The main species appear to be Fe, Ni, Cr spinel (NiFe₂O₄ or trevorite), caused by the dissolution of Monofrax E and K-3 refractory and the related increase of chromium in the glass. Laboratory glasses are typically formulated from dry chemical oxides and carbonates, and unless adjustments are made, would not reflect the changes in composition that might arise from corrosion processes in the melter. In the laboratory, SSHTM-1 and SSHTM-2 samples contained less than 0.1 vol% crystals, while production samples from the SSHTM contained from 1.48 to 4.61 vol% crystals, Table 4-7. While most of the spinel particles were small, these crystals could easily accumulate as a sludge forming layer in the melter, especially if the glass temperature is reduced or crystals are allowed to grow and settle during idling of the HTM. A spinel phase sludge layer could easily plug the bottom drain and eventually affect the electrical characteristics of the glass and electrodes.

Table 4-6. Percent of Feed Components Volatilized During Low-Level Waste Melter Tests.

Comp.	Target Glass, wt%	B&W	WSTC	USBM	Vectra	Duratek	Envitco
B ₂ O ₃	5-9	70	24	19	15	0.6	0.2
Cl	0.37	88	88	97	64	48	1 to 13
Cs ₂ O	0.15	85	84	40	41	14	0.6
F	0.31	93	91	99.7	16	53	0.9
I	0.14	95	98	N/A	83	82	10
K ₂ O	1.5-3.7	55	49	26	16	0.6	0.05
Na ₂ O	18.8-20	41	17	7.6	13.5	0.6	0.05
P ₂ O ₅	0.2	46	45	55	1.7	N/A	0.05
SO ₄	0.22	55	36	94	85	N/A	53

B&W = Babcock & Wilcox

Duratek = Duratek, Inc.

Envitco = Envitco, Inc.

USBM = U.S. Bureau of Mines

Vectra = Vectra Technologies, Inc.

WSTC = Westinghouse Science and Technology Center.

Table 4-7. Composition and Volume Percent Crystals in Small-Scale High Temperature Melter Tests 1 and 2.

Melter test	Glass	Glass comp.	Vol% crystals	Comment
Laboratory	N508 Glass	14.0 % Fe ₂ O ₃ 0.24 % Cr ₂ O ₃	< 0.1% (24 h @ 1,350 °C)	< 0.5 micron 1-4 micron spinel
SSHTM-1	Glycolic N508	12.7 % Fe ₂ O ₃ 0.93 % Cr ₂ O ₃	1.48%	spinel and second phase
SSHTM-1	Nitric N508	10.6 % Fe ₂ O ₃ 0.78 % Cr ₂ O ₃	4.59%	very small crystals
SSHTM-1 SSHTM-2	Nit+Bor N508	12.11 % Fe ₂ O ₃ 0.5 % Cr ₂ O ₃	2.98%	spinel, not evenly distributed
Laboratory	HTB65 Glass	8.6 % Fe ₂ O ₃ 0.8 % Cr ₂ O ₃	<0.01%	< 0.5 micron 0.5-5 micron spinel
SSHTM-2	DST/SST Blend (HTB65)	9.0 % Fe ₂ O ₃ 0.7 % Cr ₂ O ₃	4.61 %	spinel, varying in size

4.3.3 Radioactive Melter Tests

During Phase I of the alternate acquisition strategy (privatization) at the Hanford Site tests will be needed to evaluate the composition and properties of radioactive wastes and HLW glasses produced from these wastes. Small-scale feed preparation and crucible-level vitrification tests are necessary, and mostly sufficient, to evaluate feed rheology, offgas, feed preparation chemistry and glass properties such as viscosity, electrical conductivity, liquidus temperature and durability, and the optimum waste oxide loading of HLW glass. These tests also can be used to derive volatility data for certain radionuclides. Larger-scale tests are generally needed to assess the melt rate, foaming potential, cold cap spreading characteristics, phase separation of molten salts and refractory layers in the cold cap and volatile components in the waste. Cold simulants can be used to duplicate many of these conditions, but certain tests, especially those involving waste form qualification, must be performed with representative samples of radioactive waste. In this section, a brief summary will be provided describing radioactive melter tests at Savannah River (DWPF mini-melter), Pacific Northwest National Laboratory (radioactive liquid-fed ceramic melter), British Nuclear Fuels, Ltd (BNFL) and COGEMA (experience with radioactive wastes and simulants), and Hanford Site facilities that might be available for radioactive tests.

4.3.3.1 Savannah River Site (DWPF Mini-Melter). Small-scale tests are being conducted at Savannah River to evaluate each batch of radioactive waste before processing in the DWPF. Laboratory and bench-scale tests are currently being performed in the Shielded Cells Facility (SCF) to: (1) demonstrate compliance with the Waste Acceptance Product Specifications (WAPS), (2) verify the composition and disposition of minor and major radionuclides, and (3) demonstrate the process control model and composition control strategy for each batch of radioactive waste. The SCF includes all of the equipment needed for feed preparation and vitrification of the waste in a slurry-fed 1/100th DWPF scale joule-heated melter (DWPF mini-melter, 20 cm diameter, 15 cm deep with a 10 kg of glass inventory).

Savannah River has developed a set of algorithms or glass composition control models that can be used to assess the processability of the feed and durability of the HLW glass (feed-forward control system). These algorithms are included in the Product Composition Control System (PCCS) being used to formulate and optimize the composition of the feed to the DWPF melter.

The DWPF mini-melter was first used in 1991 to demonstrate the capabilities of the PCCS model, and later to process as much as 10 kgs of radioactive waste from tank 51 into HLW glass. Smaller batches of radioactive glass have also been made from crucible melts of pretreated tank 51 waste. These tests have generally demonstrated that HLW glasses of acceptable durability can be produced with a composition profile very similar the one predicted with the PCCS model (Andrews 1991, 1992, Ferrara 1995).

The total inventory of long-lived radionuclides must also be determined for the Savannah River glass (radionuclides with half-lives longer than 10 years and concentrations greater than 0.05 Ci up to 1,100 years after production, WAPS specification 1.2). There are twenty-four radionuclides of regulatory significance in Savannah River's waste, but only four of these will be routinely measured in the HLW glass (^{137}Cs , ^{90}Sr , ^{238}Pu , and ^{238}U). Table 4-8 provides a list of the most important radionuclides in Savannah River's waste. The minor radionuclides can be estimated from the waste dilution factors measured in the DWPF mini-melter (by comparing to concentration of minor radionuclides in waste and glass) (Bibler 1994). If this method is acceptable to the regulators, the DWPF mini-melter will be used to satisfy WAPS 1.2 reporting requirements for minor radionuclides in the waste.

Table 4-8. Principal Radionuclides in Savannah River's Waste.

Radionuclide	Half-Life	Source	Analytical Method
Ni-59	7.5E04	Neutron Act.	Seqn. & ICP-MS
Ni-63	1.0E02	Neutron Act.	Beta Count
Se-79	3.3E04	Fission Prod.	Fission Yield
Sr-90	28	Fission Prod.	Beta Count
Zr-93	1.5E06	Fission Prod.	ICP-MS
Nb-93m	16	Decay Prod.	Half-Life Calc.
Tc-99	2.1E05	Fission Prod.	ICP-MS
Pd-107	6.5E06	Fission Prod.	ICP-MS
Sn-126	1.0E05	Fission Prod.	Fission Yield
Cs-135	2.3E06	Fission Prod.	ICP-MS
Cs-137	30	Fission Prod.	Gamma Count
Sm-151	90	Fission Prod.	ICP-MS
Th-230	7.5E04	Decay Prod.	Half-Life Calc.
U-234	2.5E05	Waste Prod.	ICP-MS
U-238	4.5E09	Waste Prod.	ICP-MS
Np-237	2.1E06	Neutron Act.	ICP-MS
Pu-238	88	Neutron Act.	Alpha Count
Pu-239	2.4E04	Neutron Act.	ICP-MS
Pu-240	6.6E03	Neutron Act.	ICP-MS
Pu-241	14	Neutron Act.	ICP-MS/Gamma
Pu-242	3.8E05	Neutron Act.	ICP-MS
Am-241	4.3E02	Decay Prod.	Gamma Count
Am-243	7.4E03	Neutron Act.	ICP-MS
Cm-244	18	Neutron Act.	ICP-MS

4.3.3.2 Pacific Northwest National Laboratory (Radioactive, Liquid-Fed Ceramic Melter (RLFCM)). A pilot-scale RLFCM was installed in B-cell of the 324 Building in 1984 (Dierks 1987). This system was utilized to test radioactive feed simulants for the West Valley Demonstration Project and to produce a limited number of canisters of cesium and strontium enrich glass for the Federal Republic of Germany. The pilot-scale RLFCM consisted of a joule-heated melter, canister-handling turntable, glass-level detection system, waste feed preparation equipment, and offgas and condensate treatment racks. The melter was sized to produce about 11 kgs of glass per hour, with a melt pool area of 0.53 m², glass depth of 36 cm and nominal inventory of 475 kg of glass. Certain components in the system were designed for remote replacement, including the thermal wells, thermocouples, bubblers, the feed nozzle, air-lift lance and viewing camera (Burkholder 1985).

During the initial campaign, 300 kgs of radioactive West Valley simulant glass were produced containing 75,000 Ci of activity. Certain problems occurred, including the accumulation of cesium and sodium metaborate deposits in the melter offgas line and problems with the melter feed pump, feed nozzle, discharge heaters and tank agitator system. A new reaming device was installed to prevent future pluggage problems in the melter offgas line. Corrosion problems were also noted in the feed nozzle due to halides in the feed. Three different electrical power control schemes were evaluated for the melter, each having a different affect on the melter during glass pouring or other periods of thermal instability. These schemes included resistance feedback control (based on the electrical resistance of the glass), resistance plus glass weight factor control (to mitigate power transients following a batch pour), and glass temperature control (based on an average of three temperatures with alarms and limits). Foaming was usually observed after an upset in the process such as a glass pour or an abrupt increase in power to the electrodes. Other studies were performed to characterize melter effluents and the performance of the offgas cleaning system.

Melter feed entrainment losses were much higher than might have been anticipated from previous tests with cold simulants (volatile condensable components and particulates from the melter). The decontamination factors (DFs) for strontium and cesium were an order of magnitude lower than from cold simulant tests with comparable feeds (consistent with a large increase in particulate carryover and pluggage of the melter offgas line with cesium and sodium metaborate deposits) (Burkholder 1987). Melter operations were terminated in February, 1986 because the discharge-section heaters failed. The heaters were evidently damaged by the condensation of an electrically conductive (cesium chloride) salt layer, that caused an electrical short. This problem also seems to have been associated with uncontrolled discharges of low viscosity glass, due to foaming and pressure fluctuations in the melter. Pressure fluctuations are thought to have been caused by bridging of the cold cap and/or pluggage of the offgas line. Glass level control was also effected by continuing corrosion of the bubbler tubes (that generally lasted for only six months), and deterioration of the vacuum control system in the melter. Discharge heater failure problems were eventually solved by increasing the spacing between heater elements and filling the void with fibrous insulating material (Burkholder 1986, 1987).

4.3.3.3 British Nuclear Fuels, Ltd and COGEMA. BNFL is currently operating production-scale vitrification facilities in Sellafield, based on the COGEMA AVH and AVM designs. The main operations involve preparation of the acidic HLW feed, transfer of the HLW feed to the calciner, calcination and associated offgas treatment, melting and pouring HLW glass from the induction-heated, inconel shell melter, welding of the container lid, decontamination and monitoring of the product container, and transfer of the container to storage.

BNFL and COGEMA also have pilot-scale and full-scale non-radioactive prototype facilities in Sellafield and Marcoule. The BNFL development program is divided into four main areas: (1) engineering, including demonstration of full-scale equipment components and processes, (2) process development, comprising analytical development, simulant production and optimization of the calcination and melting processes, (3) process control, covering techniques used for product quality control, and (4) glass technology, including development of optimum glass formulations and measurement of key properties such as glass composition and durability.

Glass formulation and development is an evolutionary process. Laboratory studies are typically conducted with simulants and radioactive samples at crucible level in a glove box. Once the candidate glasses have been identified, these compositions as cold simulants are usually tested in a pilot-scale calciner and melter to optimize the process and concentration of additives used in the calciner. Pilot-scale results are then used to define the flowsheet and processes for full-scale demonstration in prototype production facilities.

With this approach, about 90 to 95 percent of the required tests can be conducted with cold simulants. Radioactive wastes as such only need to be evaluated at crucible level in the laboratory. By performing comparative tests with simulants and radioactive waste, BNFL has found that simulant results almost always match the results obtained with radioactive waste. The most surprising difference is the high volatility of certain radionuclides when radioactive wastes are used. Cesium and technetium frequently volatilize from such wastes as cesium pertechnetate (CsTcO_4). Tellurium in this waste also may alloy with Inconel, reducing the life of the induction-heated melters used by BNFL and COGEMA.

Process related measurements, such as melt rate, are determined from pilot or full-scale tests with cold simulants. Even with this level of support, it took nearly three years of effort to qualify the BNFL vitrification process, with feed-forward (glass composition) control, for the two radioactive wastes from Magnox and Thorp. These wastes are similar to the acidic, high-level PUREX wastes produced at the Hanford Site. BNFL and COGEMA HLW glasses are typically limited by heat generation or dose rate considerations to a maximum waste oxide loading of 25 to 30 wt% waste oxide, in contrast to the LTM and HTM glasses being developed at the Hanford Site where compositions are typically limited by the concentration of glass-limiting components (Fe_2O_3 , Cr_2O_3 , or P_2O_5) in the waste.

4.3.3.4 Radioactive Test Facilities at the Hanford Site. A small-scale, 1 kg/h glass melter requires about 1.5 to 2.0 L/h of feed with 10 wt% waste oxide. Small volumes of waste are typically transported between different areas of the Hanford Site using a bowling ball cask. The payload capacity of the cask is 1,130 L, that would typically provide 10 to 15 days of continuous feed for the melter. Several hot cell facilities are also available at the Hanford Site (Howden 1993). In the 324 Building, a cleanup program is currently underway to refurbish and decontaminate B-cell where the RLFCM pilot plant was located. B-cell is the largest available cell in the 324 Building (7.6 m x 6.7 m x 9.2 m high), but under current plans, this facility will be closed by October 1998.

The 325 Building high-level radiochemistry facility has three interconnecting hot cells with a total operating space of 16.2 m². This facility also has three shielded vaults with various storage tanks and pumps to transfer wastes between the vaults and hot cells. A truck port and cask handling area are also available. A-cell is the largest cell in the 325 Building (2.1 m x 4.6 m x 4.6 m high) and could be used, along with several smaller cells, for laboratory or bench-scale testing of radioactive wastes. The shielded analytical laboratory is also located in the 325 Building and includes six interconnecting hot cells (18.4 m²) and two other cells with 5.3 m² of space. Remote handling equipment is available for each cell. The 325 Building hot cell facility and equipment for testing radioactive core samples has been previously described by Morrey (1995). Other radioactive waste handling facilities may be found in B Plant, the waste encapsulation and storage facility (WESF), T Plant, the 222-S analytical laboratory, Z Plant and the unfinished fuels and materials examination facility (FMEF) in the 400 Area.

5.0 FEED PROCESSABILITY

This section provides general information on feed processability applicable to all the Hanford Site HLW. The HLW vitrification process consists of five major systems: (1) melter and turntable assembly, (2) canister-handling system, (3) feed preparation system, (4) liquid waste system, and (5) process offgas systems. The canister system has little impact on the feed processability or the performance of the melter. Melter performance characteristics were described in the previous section. In this section, glass formulation and glass composition constraints will be discussed in detail for the processing and vitrification of the Hanford Site's defense HLW.

Melter feed compositions are expected to vary due to a wide range of compositions in the tanks, incomplete recovery and mixing of wastes, and differences from tank to tank. The corresponding glass composition will therefore vary for different waste types around a nominal glass composition for each waste. Changes in glass composition affect melt viscosity, electrical conductivity, and crystallization behavior. These properties determine the processability of the glass in a joule-heated melter. Glass composition also affects chemical durability and the related acceptability of the glass for the geologic repository. The optimum glass composition has to be determined for each waste before the waste is introduced to the melter. Glass composition control is important because it is difficult to process high-viscosity glasses through the melter or to correct the composition of canister glasses with unacceptable durability properties.

5.1 WASTE FORM/WASTE FORM QUALIFICATION

The reference waste form is assumed to be borosilicate glass. This assumption is based on several reviews that have recommended borosilicate glass in the past for the low-temperature, joule-heated melter because of its durability and processability (DOE 1981 and 1990). In 1981, an independent peer review panel evaluated eight waste forms for the solidification and disposal of HLW. These waste forms were borosilicate glass, SYNROC, porous glass matrix, tailored ceramics, pyrolytic carbon and silica carbide-coated particles, concrete, metal matrices, and plasma spray coatings. Borosilicate glass was recommended based on its ability to immobilize a wide range of wastes, its insensitivity to radiation damage, and its thermal stability, chemical stability, and processability. For the HTM, sodium silicate and aluminum silicate glasses are being considered alternate waste forms based on improved WOL and chemical stability characteristics in relation to the standard borosilicate glass waste form.

Minimum requirements that waste forms must meet for acceptable geologic disposal are specified in the *Waste Acceptance System Requirement Document* (WASRD, DOE/RW-0351P, Rev. 2 to be issued March 1996) and *Waste Acceptance Product Specifications for Vitrified High Level Waste Forms* (WAPS.DOE/EM-0093, Rev. 1, May 1995). The WASRD is the senior requirements document that defines minimum

requirements and associated limits for accepting immobilized high-level waste (IHLW) product in the proposed geologic repository. The WAPS establishes the minimum set of product requirements for the IHLW product. The main difference between WASRD, Rev. 1 and the planned WASRD, Rev. 2 are: (1) DOE/Office of Civilian Radioactive Waste Management policy to accept spent HLW that does not include materials that are regulated as hazardous waste under the Resource Conservation and Recovery Act (RCRA) and (2) the concentration of plutonium should be less than 2,500 g/m³ in each canister.

The current revision of the WAPS applicable to all waste glass producers was issued by the U.S. Department of Energy-Office of Environmental Restoration and Waste Management in May 1995 (DOE 1995). The WAPS are specific instructions for the producer to assure conformance with the WASRD. The WAPS are based on repository needs and waste form and canister requirements necessary to meet those needs. The waste acceptance process requires that the DOE-OCRWM function prepare WAPS's for each waste form. Table 5-1 briefly summarizes the principal WAPS for the DWPF (as derived from 10 CFR 60 and modified to reflect current standards for product consistency, phase stability, heat source, dose rate, and subcriticality).

The WAPS criteria are important because they represent the regulatory basis for glass composition control in the vitrification process. Because the existing WAPS are based on regulatory requirements or repository design considerations, they provide useful guidance for the development of nonborosilicate or low borosilicate glass waste forms such as soda-lime or aluminum silicate glasses under development for the HTM. The WAPS compliance will generally be demonstrated by process control of the vitrification process and by direct measurement of process and product variables. Because of remote operating conditions in the plant, direct measurements will be limited only to the most important process variables such as melter feed composition and glass product quality. Glass property models and process models will be used to predict product glass composition and properties necessary to process glass through the vitrification plant. The glass product consistency criterion for DWPF, for example, requires that the mean PCT results for each waste type be at least two standard deviations below mean PCT results of the EA glass. All future testing involving the PCT will use the EA glass as an internal control, that will facilitate direct demonstration of compliance with the product consistency specification (WAPS 1.3). No melter feed will be allowed to leave the Slurry Mix Evaporator (SME) at DWPF until it has been determined, from glass property models and testing, that the melter feed will make a product that will satisfy the WAPS 1.3 specification.

Waste producers are required to document their compliance with WAPS in the Waste Form Compliance Plan (WCP), the Waste Form Qualification Report (WQR), in the production records, and in the shipping and storage records. The WCP provides general information about the waste vitrification process and product, and a detailed description of the methods and programs by which the processor will demonstrate compliance with each specification in the WAPS. The WQR is a compilation of the results of those testing and analysis programs identified in the WCP. The main objective is to confirm the producers ability to produce a product that meets specifications. Parts of the WQR may be used to

Table 5-1. Waste Acceptance Preliminary Specifications for the Defense Waste Processing Facility High-Level Waste Form.

1.1	Chemical Composition	Provide elemental and phase compositions (with ranges) for life of facility for those components >0.5 wt%
1.2	Radionuclide Inventory	Provide canister radionuclide inventories of radionuclides with half-lives longer than 10 years and concentrations >0.05 Ci up to 1,100 years after production
1.3	Product Consistency	PCTs performed for HLW to demonstrate compliance glasses with reference standard DWPF EA glass
1.4	Chemical and Phase Stability	Report glass transition temperature (T _g) and time-temperature-transformation ranges. After initial cooldown, the waste should not exceed a temperature of 400 °C during storage
2.1	Canister Material	Austenitic stainless steel
2.2	Canister Fabrication and Closure	Leaktight according to ANSI N14.5-1987 ^a
2.3	Identification and Labeling	Alphanumeric code on each canister visible
3.1	Free Liquid	None allowed in canister
3.2	Gases	None allowed in canister except helium, argon, air, and other inert. Internal pressure not to exceed 7 lb/ft ² gauge at 25 °C
3.3	Explosives, Pyrophoricity and Combustibility	None allowed
3.4	Organic Materials	None allowed
3.5	Free Volume	Less than 20 percent of canister volume
3.6	External Contamination	Less than 220 alpha dpm/100 cm ² , less than 2,200 beta-gamma dpm/100 cm ²
3.7	Heat Generation	Less than 1,500 W/canister +/- 15 percent
3.8	Minimum Dose Rate	Less than 10 ⁵ rem/h gamma and 10 rem/h neutron at surface
3.9	Chemical Compatibility	Document reactivity between waste and canister
3.10	Subcriticality	K _{eff} <0.90
3.13	Handling Features	Provide grapple and canister with lifting flange
4.0	Quality Assurance	Provide quality assurance program complying with Office of Geologic Repository/ DOE/RW-0214 ^b requirements

DWPF = Defense Waste Processing Facility

EA = Environmental assessment

PCT = Product consistency test

^aANSI, 1987, *American National Standard for Radioactive Materials--Leakage Tests on Packages for Shipment*, ANSI N14.5-1987, American National Standards Institute, New York, New York.

^bDOE-OCRWM, 1990, *Quality Assurance Requirements Document for the Civilian Radioactive Waste Management Program*, DOE/RW-0214, U.S. Department of Energy-Office of Civilian Radioactive Waste Management, Washington, D.C.

gain approval for startup, and may be used for licensing of a geologic repository for the canistered waste form.

5.2 GLASS PROPERTY CONSTRAINTS/MELTER FEED COMPOSITION CONSTRAINTS

Certain glass properties must be controlled to successfully process the glass through a joule-heated melter. Melter processing requirements have been developed for glass viscosity, electrical conductivity, and liquidus temperature. The WAPS impose limitations on glass durability and require that chemical and phase stability information must be reported. The WAPS also specifies that neither liquid-liquid phase separation nor excessively volatile or corrosive secondary phases should occur in the product glass. Table 5-2 presents the glass property constraints developed for the HWVP melter. The WAPS specification now requires the 7-day PCT test in lieu of the 28-day MCC-1 test for durability (DOE 1995).

Table 5-2. Hanford Waste Vitrification Plant Glass Property Constraints.

Component	Lower bound	Upper bound
REDOX: $\text{Fe}^{+2}/\text{Fe}^{+3}$	0.005	0.3
Viscosity: Pa-s at 1,150 °C	2	10
Electrical conductivity: S/cm at 1,150 °C	0.18	0.5
Phase behavior: liquidus temperature, °C	--	100 °C below nominal operating temperature of melter
Durability: glass release in g/m^2 for specific components during 28-day MCC-1 test	--	28

MCC = Materials Characterization Center

REDOX = Reduction oxidation.

In addition to properties necessary for control of the melter and durability of the glass, a second set of melter feed composition constraints was developed for the HWVP Project. The principal purpose of these constraints is to ensure that the melter feed compositions are restricted to the range where the glass property models are statistically valid and to provide a basis for evaluating process flowsheets, chemical additives, recycle streams, and operating conditions in the plant. To ensure that the glass product is acceptable, a CVS was conducted for the HWVP program to characterize the relationship between glass composition and glass and melt properties. The main objectives of the CVS were to: (1) develop empirical models relating the glass and melt properties to composition, (2) develop uncertainty equations for

model predictions, and (3) develop a qualified composition region for acceptable HWVP melter processability and compliance with WAPS waste form qualification requirements.

The CVS glass composition region was selected for study based on the results of a previous series of scoping and solubility studies and on projections of glass compositions that might be made from the DST wastes to be processed in the HWVP. The acceptable glass composition region is currently defined in terms of the ten major oxide components in the feed, glass additives, and recycle streams, including silica (SiO_2), boric oxide (B_2O_3), alumina (Al_2O_3), iron oxide (Fe_2O_3), zirconium oxide (ZrO_2), sodium oxide (Na_2O), lithium oxide (Li_2O), calcium oxide (CaO), magnesium oxide (MgO), and others (all remaining waste components). A total of 124 glasses were evaluated in the CVS-I and CVS-II, Phases I, II, and III, with the following properties measured: viscosity and electrical conductivity (950 to 1,250 °C), glass transition temperature, thermal expansion, crystallinity, and durability based on the MCC 28-day leach test, and the 7-day PCT. In addition, 21 other glasses were tested in CVS-II, Phase IV with only PCT and MCC-1 properties measured and 40 glasses in CVS-III, Phase I with measurements of viscosity, electrical conductivity, liquidus temperature and PCT durability. The CVS-III glasses were formulated for the high-temperature melter. Table 5-3 lists the composition ranges covered in the CVS-I and CVS-II studies and the general range over which the CVS-derived glass property models are considered to be useful and valid. It should not be assumed that all glass compositions within this region will necessarily yield an acceptable set of processing properties in the melter.

Table 5-3. Glass Composition Range Covered By Composition Variability Study (CVS-1 and CVS-2) Composition Region.

Oxide component	Lower bound (wt%)	Upper bound (wt%)
SiO_2	42	57
B_2O_3	5	20
Al_2O_3	0	17
Fe_2O_3	0.5	15
ZrO_2	0	13
Na_2O	5	20
Li_2O	1	7
CaO	0	10
MgO	0	8
Others	1	10

5.3 VISCOSITY

Industrial glasses are typically processed at a viscosity of about 10 Pa-s (100 poise) in commercial melters. This viscosity has been used for waste glass tests in pilot- and engineering-scale melters (Hagy et al. 1974, Chick et al. 1984). The viscosity of the waste glass should be maintained in the range of 2 to 10 Pa-s for optimum melter performance. Glasses with viscosity lower than about 2 Pa-s tend to penetrate into the melter refractory bricks and excessively corrode melter refractories. When glass viscosity is above about 10 Pa-s, melt rates become exceedingly low, glass becomes more difficult to pour from the melter, and such glasses may not form a consolidated monolith in the canisters....

Glass viscosity, as a function of temperature data, was modeled by: (1) the Arrhenius-type equation,

$$\ln \eta = \text{sum } (A_{Ai}X_i) + \text{sum } (B_{Ai}X_i)/T \quad (T \text{ in } ^\circ\text{K}),$$

where X_i represents the oxide mass fraction of each component and A_{Ai} and B_{Ai} are the component coefficients ($R^2 = 0.9704$); and (2) the Fulcher equation,

$$\ln \eta = \text{sum } (A_{Fi}X_i) + \text{sum } (B_{Fi}X_i)/[T - \text{sum } (T_iX_i)] \quad (T \text{ in } ^\circ\text{C}),$$

where X_i represents the oxide mass fraction of each component and A_{Fi} and B_{Fi} and T_i are the model coefficients ($R^2 = 0.9958$). The R^2 terms are a measure of the statistical fit, with $R^2 = 1.0$ representing a perfect fit of the statistical model to the experimental data. The Fulcher equation differs from the Arrhenius equation in that a temperature coefficient T_i is added to represent the temperature at which viscosity approaches infinity. The Arrhenius equation predicts a linear relationship between $\ln$ viscosity and inverse temperature. A question of practical interest is whether the simpler Arrhenius model (with 20 coefficients) adequately predicts viscosity compared to the more complicated Fulcher model (with 30 coefficients). The Fulcher model may be slightly better in the temperature range of interest for the LTM and HTM (1,150 to 1,450 $^\circ\text{C}$). Table 5-4 provides the CVS-derived Fulcher component coefficients A , B , and T_i for the first-order viscosity model.

Generally, viscosity decreases with an increase in Li_2O , Na_2O , and B_2O_3 concentration, and increases with increased SiO_2 , Al_2O_3 , and ZrO_2 content.

Table 5-4. Fulcher First-Order Viscosity Model Coefficients A, B and T_i .

Variable	Coefficient	Coefficient estimate
SiO ₂	A	-10.5899
B ₂ O ₃	A	-24.4127
Al ₂ O ₃	A	1.4998
Fe ₂ O ₃	A	-13.6326
ZrO ₂	A	-0.3590
Na ₂ O	A	2.02
Li ₂ O	A	5.4558
CaO	A	3.9535
MgO	A	5.3088
Others	A	-2.3815
SiO ₂	B	19236.3263
B ₂ O ₃	B	15922.8410
Al ₂ O ₃	B	9524.4388
Fe ₂ O ₃	B	14559.3344
ZrO ₂	B	4618.1457
Na ₂ O	B	-12965.4177
Li ₂ O	B	-39177.2042
CaO	B	-18671.4525
MgO	B	-11943.9611
Others	B	1710.2061
SiO ₂	T ₁	76.1127
B ₂ O ₃	T ₂	263.4849
Al ₂ O ₃	T ₃	178.5252
Fe ₂ O ₃	T ₄	43.6384
ZrO ₂	T ₅	540.5086
Na ₂ O	T ₆	425.7163
Li ₂ O	T ₇	474.4299
CaO	T ₈	1065.8248
MgO	T ₉	752.2421
Others	T ₁₀	270.7406

5.4 ELECTRICAL CONDUCTIVITY

In joule-heated melters, the electrical conductivity of the glass should be in the range of 0.18 to 0.5 S/cm (or $\text{ohm}^{-1}/\text{cm}$) at 1,150 °C. In the DWPF melter, the acceptable range for electrical conductivity at 1,150 °C was estimated to be 0.36 to 1.1 S/cm (Bickford et al. 1990). The current electrical conductivity range for the Hanford Site HLW melter has been extended to 0.1 to 1 S/cm at 1,150 °C (Hrma 1994). Glass must have electrical conductivity at melter operating temperatures at least two orders of magnitude higher than the refractory material in the melter walls. Such materials have never been known to have an electrical conductivity of more than 0.001 S/cm at 1,150.°C...Thus, 0.1 S/cm appears to be a safe and acceptable lower limit for glass electrical conductivity. The upper limit of 1.0 S/cm is an order of magnitude higher than the lower limit and in reasonably good agreement with the DWPF electrical conductivity estimate. If the electrical conductivity of the glass is too low, electrical shorting could occur through the melter's refractory, and larger power supply systems could be required due to higher voltages across the electrodes. High-electrical conductivities could result in excessively high-current densities across the electrodes and accelerated corrosion due to localized overheating of the electrodes.

Electrical conductivity, as a function of temperature data, was modeled as a simultaneous function of temperature and composition by expressing the Arrhenius equation coefficients as a first-order mixture model:

$$\ln E = \text{sum } (A_i X_i + B_i X_i / T) \quad (T \text{ in } ^\circ\text{K}),$$

where X_i represents the oxide mass fraction of each component, and A_i and B_i are component coefficients ($R^2 = 0.9582$). Table 5-5 provides electrical conductivity Arrhenius equation coefficients A and B.

Electrical conductivity is primarily affected by the concentration of alkaline oxides such as Li_2O and Na_2O in the glass while other components have little or no effect. Both Li_2O and Na_2O increase electrical conductivity, Li_2O considerably more than Na_2O .

5.5 LIQUIDUS TEMPERATURE

Feed components that are not completely soluble in glass may cause melter operating problems and reduce the operable life of the melter. Insoluble or crystalline solids that are more dense than the melt could settle and produce a sludge layer on the floor of the melter. If the sludge is electrically conductive, current flow through the sludge could rapidly corrode electrodes and substantially reduce current densities and melting capacity. Solids could also clog the pour spout or increase the viscosity of the glass.

Table 5-5. Arrhenius Equation Electrical Conductivity Coefficients A and B.

Variable	Coefficient	Coefficient estimate
SiO ₂	A	8.12
B ₂ O ₃	A	12.82
Al ₂ O ₃	A	7.14
Fe ₂ O ₃	A	9.94
ZrO ₂	A	7.93
Na ₂ O	A	6.05
Li ₂ O	A	7.47
CaO	A	14.41
MgO	A	10.39
Others	A	18.11
SiO ₂	B	-10283
B ₂ O ₃	B	-15135
Al ₂ O ₃	B	-8227.18
Fe ₂ O ₃	B	-10608
ZrO ₂	B	-9723.86
Na ₂ O	B	7089.48
Li ₂ O	B	22484
CaO	B	-18769
MgO	B	-13414
Others	B	-20653

The temperature where the first crystals precipitate in the glass is called the liquidus temperature (T_L). The HWVP Project adopted the DWPF criteria for T_L , that specifies an upper limit of 1,050 °C. This is 100 °C below the nominal operations temperature of the bulk glass and is intended to prevent precipitation of solids in the cooler zones of the melt pool. Only trace components such as insoluble noble metal oxides should be present in the melter if the glass is formulated to meet the liquidus temperature (T_L) constraint. Liquidus temperature considerations are important because liquidus temperature is currently the limiting constraint for several of the Phase I feeds.

Liquidus temperature can be determined by heat-treating either crystal-free or pre-crystallized samples of glass in a gradient-temperature furnace. An alternate method is to anneal the glass at 1,050 °C for 24 hours to induce crystallization at the liquidus temperature limit and from this determine whether the actual liquidus temperature is above or below this threshold.

Useful information also can be derived from canister centerline cooling (CCC) tests in which samples are taken through a predetermined cooling profile to replicate thermal conditions at the canister centerline. Crystals formed during CCC tests provide insight as to the likely phases that would form during cooling in the canister. Crystalline phases in the glass are characterized by optical microscopy with transmitted and reflected light, X-ray diffraction, and scanning electron microscopy with energy dispersive spectroscopy (SEM/EDS) to identify crystalline phases and to determine semiquantitatively the volume fraction of such crystals.

The major crystalline phases identified during the CVS were spinel phases (iron, nickel, chromium), lithium silicate, clinopyroxene (calcium, iron, and magnesium silicates), zircon (zirconium silicate), silica, nepheline (sodium aluminum silicate) and eucryptite (lithium aluminum silicate). Noble metals, mostly RuO₂, were present as undissolved particles and sometimes as small needles with metallic particles of Ru/Rh surrounded by layers of Pd/Ag/Te alloy. In some glasses, spinel phases and other crystals were nucleated on RuO₂.

Spinel structures are represented by the formula $M^{II}M^{III}_2O_4$, where M^{II} consists of di-valent Fe, Mg, or Mn and M^{III} tri-valent Fe, Cr, or Al.

The liquidus temperatures for spinel, clinopyroxene, and zirconium-containing crystals were statistically fit to a limited number of CVS glasses, using a first-order mixture model of the form:

$$T_L = \sum (B_i X_i)$$

where: X_i represents the mass fraction of each major oxide in the glass

B_i is the component coefficient.

Table 5-6 summarizes the first-order model coefficients and range of measured liquidus temperatures for CVS glasses (Kim and Hrama 1995).

Table 5-6. Component Coefficients for First-Order Model of Liquidus Temperature.

Component	Clinopyroxene	Spinel phases	Zirconium crystals
SiO ₂	855.65	989.31	753.78
B ₂ O ₃	314.72	666.42	1095.83
Al ₂ O ₃	1319.48	1735.03	1138.06
Fe ₂ O ₃	1506.69	2256.00	1461.04
ZrO ₂	1844.50	928.11	4541.99
Na ₂ O	38.83	3.77	74.31
Li ₂ O	-207.05	-128.77	-956.39
CaO	1372.44	1366.21	886.76
MgO	2387.62	2830.58	2458.47
Others	1357.40	1005.56	657.99
R ²	0.905	0.643	0.79
Measured T _l range	761 to 969 °C	800 to 1,129 °C	856 to 1,129 °C

The clinopyroxene model provides a good statistical fit to the experimental data, but this model is relatively unimportant because clinopyroxene crystals form at low temperatures and as such are unlikely to affect melter performance. For spinel phase crystals, the statistical fit is relatively poor based on the 28 CVS glasses used to develop this model. The fit for the liquidus temperature of zirconium-containing crystals is reasonably good ($R^2 = 0.79$), although three different crystals are included in the model (zircon, sodium-zirconium silicate, and zirconium oxide). A better fit might be obtained if a sufficient amount of data were available to develop a separate model for each crystal form. The zirconium model also may have some predictive capability at higher temperatures because the model predicts, for all six CVS glasses in which zirconium crystals were found at the maximum operating temperature limit of the gradient temperature furnace (1,129 °C), liquidus temperatures above 1,129 °C. Judging from the size of the coefficients in the zirconium and spinel models, alkali oxides (Li₂O and Na₂O) should depress the liquidus temperature with Li₂O having the strongest effect. Boric oxide, CaO, and Al₂O₃ should have little effect on the liquidus temperature of zirconium-containing glasses while ZrO, MgO, and Fe₂O₃ may have a significant effect on such glasses.

Recently, the predicted trends for B_2O_3 , Na_2O , and Li_2O were experimentally confirmed for spinel phases and zirconium crystals in high-waste-oxide-loaded NCAW glasses (Kim and Hrma 1994). Figure 5-1 shows the comparison between predicted liquidus temperature of the zirconium-containing phase and measured liquidus temperatures and primary phases in CVS glasses. Figure 5-2 depicts a similar comparison between measured and predicted liquidus temperature of spinel. In CVS-I and II, only 10 of 124 glasses were found to have spinel liquidus temperatures above $1,050^\circ C$. Nine of these glasses had more than 10 wt% Fe_2O_3 , which suggests that iron has a significant effect on the precipitation of spinel in borosilicate glass. Twenty-three of 40 CVS-III glasses had spinel liquidus temperatures above $1,050^\circ C$. All of these glasses were very low boron or non-borosilicate glasses, which suggests that boron also has a large effect on the liquidus temperature of spinel.

Figure 5-1. Predicted Versus Measured Liquidus Temperature of Zr-Containing Crystals for the First-Order Mixture Model.

Figure 1 is a scatter plot showing the relationship between Predicted Liquidus Temperature (°C) on the y-axis and Measured Liquidus Temperature (°C) on the x-axis for the FeO-Fe₂O₃ system. The plot includes data points for spinel phase (open circles) and T_L (Measured) > T (Max, tested) (filled triangles). A solid line represents the linear regression fit, with R² = 0.643. The x-axis ranges from 800 to 1200 °C, and the y-axis ranges from 800 to 1200 °C.

Spinel in HLW glasses normally contain iron, chromium, and nickel associated as elemental oxides in magnetite (Fe_3O_4), chromite (FeCr_2O_4), and trevorite (NiFe_2O_4). Magnetite is the spinel phase most commonly found in CVS glasses with nickel and chromium as minor components. Melter operating temperatures and REDOX conditions have a strong thermodynamic influence on the form of iron in the melt (magnetite or hematite, Fe_2O_3). Crystallization is a kinetically dependent process in that some crystalline phases may be thermodynamically favored but may not be present because their crystallization rate is too low. Spinel phases crystallize rapidly in glass but dissolves very slowly after they precipitate.

Another approach developed for liquidus temperature prediction is the phase-equilibria liquidus temperature model (Ecole Polytechnique). This model uses a set of equations describing the thermodynamic properties of glass as functions of temperature and composition based on phase diagrams and thermodynamic data for binary and ternary systems. A quasichemical model (QCM) was developed to represent the thermodynamic properties of the glass as a function of composition and temperature. The QCM can be used to predict the Gibbs free energy of the multicomponent liquid using optimized parameters derived from available phase data. The output of the model includes T_L and the primary crystalline phase, equilibrium crystallization at temperatures below T_L (including fractions and activities of all crystalline phases considered) and metastable T_L with suppression of crystallization of the primary phase. For a given glass, the model predicts one or more possible primary phases with the corresponding liquidus temperature. The first phase may not be present in the glass because of slow kinetics (slow crystallization rate or low concentration of components in the melt). The experimentally derived primary phase agreed with the predicted first phase in 31 of 89 CVS glasses tested, agreed with the predicted second or third phase in 20 glasses, and showed no agreement in the remaining 38 glasses. When the measured versus predicted liquidus temperatures were compared for the 89 CVS glasses, the statistical fit was found to be very poor ($R_2 = 0.431$). For spinel liquidus temperature prediction, agreement was satisfactory in some cases and poor in others. When measured versus predicted liquidus temperatures for spinel phases were compared, the average temperature deviation was 68 °C.

Still another model is the liquidus temperature spinel model developed for SRS glasses (Jantzen 1991). This model is based on the free energy of formation of the liquidus phases such as spinel and nepheline. The model assumes that the liquidus temperature of spinel is proportional to a pseudo-equilibrium constant given by the following equation:

$$k = \frac{m(\text{Fe}_2\text{O}_3) \Delta G_{\text{fm}}(\text{NiFe}_2\text{O}_4)}{m(\text{SiO}_2) \Delta G_{\text{fm}}(\text{SiO}_2) - m(\text{Al}_2\text{O}_3) \Delta G_{\text{fm}}(\text{NaAlSiO}_4)}$$

where: $m(\text{Fe}_2\text{O}_3)$, $m(\text{SiO}_2)$, and $m(\text{Al}_2\text{O}_3)$ are mole fractions

ΔG_{fm} is the free energy of formation of $(\text{NiFe}_2\text{O}_4)$ -134 kcal/mole, (SiO_2) -156 kcal/mole, and (NaAlSiO_4) -360 kcal/mole.

For SRS glasses, $R_2 = 0.77$, when $T_L = A + BK$, $A = 803.6$ °C, and $B = 2277$ °C. When this model is applied to CVS glasses, the statistical fit is very poor ($R^2 = 0.178$) partially because CVS glasses were not used to develop this model. It appears this model is not useful for glasses with large variations in major component concentration such as those from the Hanford Site.

5.5.1 Model Refinements During 1996

New models were developed in 1996 for estimating the liquidus temperature of spinel. These models are derived from liquidus temperature measurements of 61 CVS glasses, 19 Savannah River glasses and 11 Phase I glasses (see Tables 6-2, 6-3, and 6-4 for the estimated composition and properties of the Phase I glasses). The liquidus temperature of spinel was statistically fit to first and second-order mixture models of the form:

$$T_L = \sum (b_i X_{y_i}) \quad \text{First-order model } (R^2 = 0.845)$$

$$T_L = \sum (c_i X_{y_i}) + \sum (d_i X_{y_i} X_{z_i}) \quad \text{Second-order model } (R^2 = 0.9488)$$

where: T_L is the liquidus temperature in °C.

y_i and z_i represent the mass fractions of each major oxide and minor oxides such as Cr_2O_3 and NiO in the glass.

b_i , c_i and d_i are single and mixed component coefficients for the first and second order terms.

Table 5-7 provides the statistically derived component coefficients for the first and second-order models, while Table 5-8 compares the measured and estimated liquidus temperatures for the Phase I glasses and a selected group of CVS and Savannah River glasses with the poorest fit based on the first-order model (more than 50 °C above or below the measured T_L).

The root mean square deviation between predicted and measured T_L for this group of samples is 53 °C for the first-order model and 25.5 °C for the second-order model. Predictive performance can be evaluated by comparing these models to glasses that were not used to develop the models. PNNL is currently developing another set of 36 glasses from which they will obtain liquidus temperature measurements for testing validity of current models. These glasses are of special interest because they were formulated around a spinel-forming baseline composition (similar to the Phase I nom-3 glass), with targeted variations in Al_2O_3 , B_2O_3 , Cr_2O_3 , Fe_2O_3 , Li_2O , MgO , MnO , Na_2O , and SiO_2 to measure the effect of each component on the liquidus temperature of spinel. Results of this investigation may not be available until the end of FY 1996. This study is important because it is the first study to systematically evaluate the effects of chromium, nickel and manganese.

Spinel phase and zirconium oxides and silicates are the crystalline phases most often observed in CVS glasses at high temperature (1,050 °C). The best liquidus temperature models appear to be the CVS-derived, first-order model for zirconium-containing crystals, and the statistically-derived second-order model for spinel. Spinel is clearly the dominate phase of concern in the melter because of low zirconium concentration in the Phase I feeds. However, nepheline, and potentially eucryptite ($LiAlSi_2O_6$), are the most important phases in the canistered glass. The precipitation of nepheline during canister centerline cooling is commonly associated with poor durability results from the PCT test. The second-order spinel model, however, is especially important because this model allows one to predict the crystallization behavior of glasses in the melter. This spinel model, together with other property models such as the ones for viscosity, durability, and nepheline can be used to estimate the maximum WOL and limiting glass composition of the Phase I feeds.

Table 5-7. First and Second-Order Spinel Liquidus Temperature Model
Coefficients B, C, and D.

Variable	Coefficient	First-order model	Coefficient	Second-order model
SiO ₂	B	1247.477	C	-3252.578
B ₂ O ₃	B	266.961	C	1913.452
Na ₂ O	B	- 654.105	C	671.302
Li ₂ O	B	-2004.931	C	-778.785
CaO	B	920.331	C	3921.812
MgO	B	1947.495	C	-3978.842
Fe ₂ O ₃	B	2118.935	C	2206.712
Al ₂ O ₃	B	2084.204	C	3052.081
ZrO ₂	B	2672.485	C	1934.87
Bi ₂ O ₃	B	876.592	C	2331.235
Cr ₂ O ₃	B	14956.243	C	37497.11
MnO	B	1092.062	C	23637.442
NiO	B	4463.781	C	3531.782
P ₂ O ₅	B	1006.421	C	3116.844
Others	B	911.304	C	1971.079
SiO ₂ xSiO ₂			D	6325.440
CaOxCaO			D	-16731.47
MgOxB ₂ O ₃			D	26612.647
MgOxLi ₂ O			D	82967.224
Fe ₂ O ₃ xFe ₂ O ₃			D	8336.89
ZrO ₂ xAl ₂ O ₃			D	36523.126
Cr ₂ O ₃ xCr ₂ O ₃			D	-1.584E6
MnOxSiO ₂			D	-38271.46
R ²	0.845139		0.948884	
R ² (adjusted)	0.816612		0.932346	

Table 5-8. Comparison of Measured to Predicted Spinel Liquidus Temperatures for Phase I Glasses and Selected Group of CVS and Savannah River Glasses.

Glass	T _L measured, °C	T _L predicted, °C first-order model	T _L predicted, °C second-order model
Nom-2*	1,009	983	990
Nom-3*	1,047	1,017	1,053
Nomc-1*	1,005	1,047	1,042
Nomc-2*	998	996	998
C106A-2*	1,004	1,035	1,011
C106A-3*	1,029	999	979
C106A-4*	885	943	927
C106B-1*	986	941	999
C106B-2*	894	902	877
AZ-3*	1,048	1,017	1,026
AZ-5*	953	971	956
CVS1-19	961	1,017	989
CVS1-20	966	1,017	988
CVS2-30	1,090	1,017	1,052
CVS2-42	1,093	1,020	1,060
CVS2-46	1,004	926	1,009
CVS2-51	899	971	921
CVS2-56	1,117	1,039	1,110
CVS2-92	905	852	916
CVS2-97	887	971	921
CVS3-9	1,374	1,308	1,348
CVS3-13	1,295	1,242	1,266
CVS3-18	1,293	1,238	1,310
CVS3-24	1,127	1,092	1,118
AH-168AL	897	819	864

*Phase I glass described in Tables 6-2, 6-3, and 6-4.

5.6 DURABILITY

The durability of borosilicate glass is the property defining radionuclide release from the waste form. The intrusion of groundwater into and through a geologic repository is the most likely mechanism for transporting radionuclides into the biosphere. Thus, it is important that nuclear waste glasses be stable in the presence of groundwater over the geologic time scale.

The Product Consistency Test (PCT) has been developed for assessing the product acceptability in relation to the selected standard EA glass. The PCT, which uses a crushed sample of glass, is sensitive to glass composition and homogeneity and can be performed in a remote environment with radioactive or hazardous glasses because of simple sample preparation and test procedures. The test can be performed in only 7 days. Crushed glass, preferably of 100 to 200 mesh at the recommended ratio of 10 ml of solution to 1 g of glass, is immersed in American Society for Testing and Materials Type-1 water for a minimum of 7 days at 90 °C. Leachates are filtered and analyzed for Ph and elemental concentrations of interest. Results are reported as normalized elemental releases of

$$r_i = m_i / (F_i A)$$

where: r_i is the mass release of element i (g/m²)

m_i is the mass of the i -th element in solution (g)

F_i is the i -th element mass fraction in glass

A is the sample surface area (m²).

The WAPS was revised in 1993 and once again in 1995 (DOE 1995) and now requires the normalized release of boron, lithium, and sodium determined by the PCT to be less than the release of these elements from the DWPF EA glass. Normalized boron, lithium, and sodium release values for EA glass have been determined and reported by DWPF (*Technical Bases for the DWPF Glass Product Control Program*, WSRC-IM-91-116-5, Rev. 1, Plodinec 1995). The EA glass dissolution rates for these elements are shown in Table 5-9. At DWPF, the product acceptance standards require that the mean PCT results for each waste type be at least two standard deviations below the mean PCT results of the EA glass. All future testing involving the PCT will use the EA glass as an internal control that will facilitate direct demonstration of compliance with the product consistency specification (WAPS 1.3).

The effect of crystallization on high-aluminum (5.6 and 11.1 wt%) glass and high-iron borosilicate glass was studied using the MCC-I and Corning¹ Glass Works tests (Jantzen and Bickford 1985). These studies concluded that formation of a Ni, Fe, spinel phase

¹Corning is a trademark of Corning Glass Works.

(NiFe_2O_4) had little or no effect on leachability and the formation of 10 to 20 vol% acmite ($\text{Na}_2\text{O} \cdot \text{Fe}_2\text{O}_3 \cdot 4\text{SiO}_2$) produced a 12 to 23 percent increase in glass dissolution as measured by PCT (Jantzen 1984). When MCC-1 tests were conducted on a glass of lower alumina content (SRL-165 glass with high iron, 2.2 wt% Al_2O_3 and 6.5 wt% B_2O_3), boron releases were found to increase by a factor of 10 after heat treatment (crystallization) (Jantzen and Bickford 1985). This may indicate the possible crystallization of nepheline ($\text{NaAlSi}_3\text{O}_6$) in the Savannah River 165 glass, that would lead to the depletion of aluminum and the loss of durability in the matrix glass. These effects have been observed with certain CVS glasses (CVS2-63, CVS2-33). The precipitation of eucryptite ($\text{LiAlSi}_2\text{O}_6$) also reduces the amount of aluminum in the matrix glass and (Exnar 1995). Eucryptite crystals have previously precipitated from borosilicate glass with 16.4 to 24.6 wt% Al_2O_3 and 5.9 to 6.5 wt% Li_2O .

Table 5-9. Environmental Assessment Glass Dissolution Rates.*

Property	Normalized Release, g/m^2	Standard Deviation, g/m^2
Boron	8.35	0.61
Lithium	4.785	0.37
Sodium	6.675	0.45

*In deionized water, product consistency test (7 day, $A/V = 2000 \text{ m}^{-1}$). Averages and standard deviations are pooled values from six data sets.

Figure 5-3 compares the normalized boron release from quenched and canister centerline cooled (CCC) glasses in the CVS database. This comparison is important because it shows the effect of crystallization on the durability of the canister centerline glass. The effect of crystallization varies with the type and volume percent crystals in the glass. The precipitation of nepheline, cristobalite (SiO_2) or eucryptite ($\text{LiAlSi}_2\text{O}_6$) during CCC cooling, causes a substantial increase in normalized boron release (compared to quenched samples of the same composition). For specific compositions of interest, one should refer to CVS2-16, CVS2-33, CVS2-35, CVS2-63, and CVS2-85 for nepheline precipitation, CVS2-78, CVS2-79 and CVS2-80 for cristobalite, and CVS2-68 and CVS2-80 for eucryptite. Other crystals exert a minor to moderate effect on durability, including hematite (Fe_2O_3), clinopyroxene and zircon, while lithium silicate (Li_2SiO_3) actually causes a decrease in boron release (increase in durability) due to the removal of lithium from the matrix glass. The adverse affects of nepheline and eucryptite precipitation are caused, as stated earlier, by the removal of aluminum from the matrix glass.

Figure 5-3. Comparison of Normalized Boron Release From Quenched and Canister Centerline Cooled Glass.

As part of the FY 1996 glass work, PNNL evaluated potential Phase I glass compositions for impact of nepheline precipitation during glass cooling. PNNL compared nepheline precipitation data from the relatively complex HLW glasses with the published ternary phase diagram for the $\text{Na}_2\text{O}-\text{Si}_2\text{O}-\text{Al}_2\text{O}_3$ system.

The phase diagram for the $\text{Na}_2\text{O}-\text{Al}_2\text{O}_3-\text{SiO}_2$ system is shown in Figure 5-4. The central area of this diagram depicts the composition region where nepheline is expected to form. Figure 5-5 provides a plot of the normalized canister centerline cooled glass compositions where Na_2O , Al_2O_3 and SiO_2 have been normalized to the sum of the three components. The open circles represent glasses that formed nepheline (in quenched or canister centerline cooled samples), while the closed points represent glasses that did not form nepheline during CCC cooling. In almost every case, the nepheline points reside in the nepheline designated area of the phase diagram. Only 1 of 27 nepheline forming glasses (CVS3-16) resides outside the nepheline area, while only 2 of 157 CVS and Phase I glasses (including C106B-1) that did not form nepheline reside inside of the nepheline area. The composition of the CVS3-16 glass was altered by the precipitation of corundum (Al_2O_3) that changed the matrix composition into one that rests in the nepheline region.

Studies are currently underway to map the nepheline composition region with 13 newly formulated glasses and to reevaluate the nepheline forming behavior of the Savannah River PUREX glass. Since many of these glasses have different melting temperatures, the canister centerline cooling curves are also being adjusted to compensate for these differences. The nepheline region in the phase diagram is approximately defined by the following limits:

1. $0.43 < \text{SiO}_2 < 0.62$
2. $\text{Al}_2\text{O}_3 < 1.67 * \text{Na}_2\text{O}$
3. $\text{Al}_2\text{O}_3 < 0.28$
4. $\text{Al}_2\text{O}_3 + 0.57 * \text{SiO}_2 > 0.43$

where Na_2O , Al_2O_3 and SiO_2 are mass fractions of each component normalized to the sum of the three components. The nepheline region can be avoided by defining a minimum limit for SiO_2 such that glasses with a normalized SiO_2 content ≥ 0.62 should not be susceptible to nepheline precipitation in the canister. Based on recent tests, the minimum SiO_2 limit may be reduced to 0.6 as more progress is made in determining the composition ranges where nepheline precipitation could be a problem. The defined limits for nepheline are especially important for certain wastes, such as Tank 241-C-106 waste, where the WOL limits are likely to be determined by the nepheline forming characteristics of the glass.

Figure 5-4. $\text{Na}_2\text{O} - \text{Al}_2\text{O}_3 - \text{SiO}_2$ Phase Diagram

Figure 5-5. Nepheline Region in Na_2O - Al_2O_3 - SiO_2 Phase Diagram

It is not clear if nepheline laden glasses would be acceptable to the repository because of the significant inventory of crystals and nonuniform distribution of properties in the canistered glass. Also with slower annealing, nepheline laden glasses could easily fail the PCT durability criteria. During the DWPF Waste Qualification Runs, the maximum amount of crystallinity was found to be 5.6 vol% (1.9 vol% spinel and 3.7 vol% acmite- $\text{NaFeSi}_2\text{O}_6$) (WP-14, canister S00155, WSRC-IM-91-116-5, Plodinec et al. 1996). The PCT results for maximum crystallinity samples were within 5 to 6 percent of the results for quenched samples from the same canister height, thus demonstrating the general uniformity of the DWPF glass (Table 11, WSRC-IM-91-116-5). The waste producer is currently required to report all of the crystalline phases in the glass (WAPS 1.1) and to determine the effect of any phase changes, from the WAPS 1.1 baseline conditions, on the PCT response of the borosilicate glass (WAPS 1.4). However, the WAPS criteria do not specifically address crystallinity limits in the glass, in terms of induced stress limits or performance assessment characteristics of the crystal laden glass. Nor do the WAPS criteria provide any guidance as to the proper course to take in formulating high-waste-oxide-loaded glasses that may produce crystals in the melter or during canister cooling.

In Figure 5-6, normalized sodium and silica releases are shown as a function of NCAW WOL. This figure is instructive because it shows that the release of sodium does not increase until NCAW WOLs approach about 70 wt% in glass. Figure 5-7 illustrates the effect of aluminum oxide on normalized boron release from CVS glasses and the relative effect of changing the aluminum oxide content in the standard HW39-4 glass. This correlation shows that aluminum oxide has a significant effect on PCT (short-term) durability in that normalized boron releases are reduced as aluminum oxide content is raised in borosilicate glasses. Generally, it has been found that glass durabilities increase with increasing concentrations of Al_2O_3 , ZrO_2 , and SiO_2 in glass (with aluminum oxide having the dominate effect) and decrease with increasing concentrations of Li_2O , Na_2O , B_2O_3 , and MgO .

These roles can be explained by considering structural effects in glass. The addition of Na_2O interrupts the silica-oxygen-silica bonds that form the glass network by converting these bonds into relatively weak non-bridging oxygen (NBO) bonds. When Al_2O_3 is added, the durability of the glass is improved because the NBO bonds are converted into stronger aluminum to oxygen bonds. Other components such as ZrO_2 , Fe_2O_3 , and B_2O_3 also improve the durability of glass by reducing the number of NBO bonds (with the relative binding strength to alkalis being $\text{Al}_2\text{O}_3 > \text{ZrO}_2 > \text{Fe}_2\text{O}_3 > \text{B}_2\text{O}_3 > \text{CaO}$). Structurally, Al_2O_3 , ZrO_2 , Fe_2O_3 , B_2O_3 , and SiO_2 are considered network formers and Na_2O , Li_2O , K_2O , Rb_2O , and Cs_2O network breakers, with several of these components having an ambivalent role depending on the relative concentration of NBO bonds in the glass network. For LLW glasses (with 20 wt% Na_2O), the most durable compositions contain from 9 to 12 wt% Al_2O_3 and 6 to 9 wt% B_2O_3 (Kim 1995).

Figure 5-6. Effect of Neutralized Current Acid Waste Loading on 7-Day Product Consistency Test Normalized Element Sodium and Silicon Releases.

Figure 5-7. Effect of Aluminum Oxide on Normalized Boron Release from Composition Variability Study and HW39-4 Glasses.

Second-order mixture models of the form $\ln r = \sum a_i X y_i + \sum c_i X y_i X z_i$, where a_i , c_i and y_i , z_i are the single and mixed component coefficients and mass fractions, respectively, were fitted to the natural logs of the averaged silicon, boron, lithium, and sodium elemental release data from CVS-I and CVS-II, Phases I, II, and III, glasses. The elemental release coefficients for second-order model #3 are shown in Table 5-10 based on the results of a 7-day PCT.

Judging from the relative value of the coefficients, alumina, sodium, and lithium oxides seem to have a dominant effect on durability properties of borosilicate glass. For nonborosilicate glass compositions developed for the HTM, durability is less important as a composition constraint because high-temperature glasses are usually much more durable than the reference EA glass.

5.7 GLASS SOLUBILITY CONSTRAINTS

Some minor components are relatively insoluble in borosilicate glass. The most important of these from a waste composition perspective are chromium oxide and phosphate. Other minor components such as chloride, fluoride, and sulfate also have very low solubility in HLW glass. Uranium and bismuth are known to be present in significant concentration in SST wastes, but the solubility limit and component interaction effects of these components have not been fully determined in borosilicate glass. Noble metals (including silver and tellurium) in the Phase I wastes are expected to be a concern in flat-bottom melters and should be limited to ensure an acceptable melter operating life. Titanium oxides may be important because such materials may be used as cesium absorbers to remove cesium from the LLW stream. These materials with captured cesium can be added to HLW feed for ultimate disposal. In this section, minor component solubility limits for HLW glass will be discussed. These limits are expected to have a significant impact on glass composition, maximum WOL constraints for certain wastes, and melter operations.

5.7.1 Chromium Oxide Solubility

Chromium^{III} and Cr^{VI} are both present in borosilicate glass, with Cr^{III} being the dominant and most stable oxidation state of chromium. Chromium^{III} oxide has poor solubility in borosilicate glass because of its high-melting temperature (1,900 °C) and poor miscibility with SiO₂. The solubility of Cr₂O₃ increases with increasing temperature and increasing alkali metal content in the glass (Volf 1984). In most commercial glasses, the Cr₂O₃ solubility limit is about 2 wt% based on physical property constraints of the glass. If the ratio of Cr^{VI}/Cr^{III} in the glass melt increases to 0.01, alkaline chromates tend to separate from the glass. Under these conditions, the alkaline chromates will concentrate in a yellow surface layer where they are relatively soluble leading to poor durability characteristics in the glass (Sussmilch and Jouan 1993). Chromates are preferentially formed in glasses with high-alkali-metal content (basic glasses). Because of REDOX considerations, the amount of Cr⁶⁺ in a molten glass should be negligibly small at temperatures greater than 1,000 °C.

Table 5-10. Coefficients for Second-Order Model (#3) of Natural
Logarithm of Average 7-Day Product Consistency Test
Normalized Elemental Releases (g/m^2).

Component	PCT silicon	PCT boron	PCT lithium	PCT sodium
SiO_2	-2.3415	-4.1267	-3.3159	-1.7652
B_2O_3	2.2959	-2.7803	12.4446	-10.4721
Al_2O_3	-28.9796	-39.6897	-37.6244	-32.6424
Fe_2O_3	-5.6296	-0.7342	-5.5866	-2.8512
ZrO_2	-17.2431	-21.8129	-10.3597	16.1412
Na_2O	17.8263	19.7648	16.3850	12.5007
Li_2O	18.0258	25.1279	16.9458	7.5967
CaO	11.2689	7.8944	20.5631	8.5246
MgO	-1.7491	-51.2479	12.1879	-17.0361
Others	-2.5487	4.3558	-19.0889	0.7069
$\text{Al}_2\text{O}_3 \times \text{Al}_2\text{O}_3$	96.5647	105.2815	99.7873	89.9973
$\text{SiO}_2 \times \text{MgO}$		119.5209		57.6768
$\text{Na}_2\text{O} \times \text{ZrO}_2$		70.4225		
$\text{CaO} \times \text{ZrO}_2$	95.2066	101.8736		
$\text{B}_2\text{O}_3 \times \text{CaO}$		-80.9291	-119.8254	-96.6209
$\text{Na}_2\text{O} \times \text{CaO}$		-90.8996	-120.7020	
$\text{MgO} \times \text{ZrO}_2$	109.7168	146.7060		
$\text{Na}_2\text{O} \times \text{Al}_2\text{O}_3$	-53.2773			
$\text{B}_2\text{O}_3 \times \text{Na}_2\text{O}$	-40.6487			
$\text{SiO}_2 \times \text{CaO}$	-43.2976			
$\text{Li}_2\text{O} \times \text{MgO}$			165.6870	
$\text{MgO} \times \text{Al}_2\text{O}_3$			-153.5615	
$\text{Fe}_2\text{O}_3 \times \text{Al}_2\text{O}_3$			82.5595	
$\text{Na}_2\text{O} \times \text{Li}_2\text{O}$				152.3524
$\text{SiO}_2 \times \text{ZrO}_2$				-53.2743
$\text{Li}_2\text{O} \times \text{Al}_2\text{O}_3$				-86.3851
$\text{B}_2\text{O}_3 \times \text{B}_2\text{O}_3$		76.5449		94.9874
R^2	0.8471	0.9106	0.9058	0.9238

PCT = Product consistency test.

Also, there does not appear to be any evidence of chromate molten salt segregation in chromium rich, low-level waste glasses formed at 1,150 and 1,450 °C (Li 1995).

A single-component study was recently performed to assess the effect of adding chromium (0 to 2 wt% Cr_2O_3) to HW39 reference borosilicate glass (with 11.1 wt% Fe_2O_3) (Bates and Bowen 1987). In general, chromium was found to have only a very minor effect on viscosity and electrical conductivity, and it had no material effect on the physical or leachability properties of the glass. The only significant effect was an increase in the amount of crystalline material in the melt in direct proportion to the Cr_2O_3 concentration above 0.5 wt%. Crystalline material was not observed in the glass with 0 wt% Cr_2O_3 . As the Cr_2O_3 concentration increased from 0.5 to 2.0 wt%, an appreciable amount of crystalline material (1 to 5 wt%) was formed. The Cr_2O_3 solubility limit appears to be about 0.5 wt% in HW39 glass (at 1,150 °C), consistent with earlier HWVP, DWPF, and WVDP studies of other borosilicate glasses. Cubic Fe, Cr spinel (FeCr_2O_4) crystals, about 5 μ in diameter, were the only crystals observed above 1.0 wt% Cr_2O_3 . Between 0.5 and 1.0 wt% Cr_2O_3 platelets of the sesquioxide ($\text{FeCr})_2\text{O}_3$ of about 35 μ in size were also present with spinel in the glass.

In the CVS, slightly different results were obtained. Chromium oxide crystals with sharp facets were observed after the canister centerline cooling test in one sample with 0.89 wt% Cr_2O_3 (CVS2-55) but not in other samples with up to 0.72 wt% Cr_2O_3 (CVS2-56 and -57). These samples also contained only 7.33 wt% Fe_2O_3 , that might explain the somewhat higher Cr_2O_3 solubility limit in these glasses compared to HW39 glass (11.1 wt% Fe_2O_3). Spinel phases have only rarely precipitated in CVS borosilicate glasses with less than 10 wt% Fe_2O_3 at temperatures above 1,050 °C. The sample with 0.89 wt% Cr_2O_3 had a liquidus temperature of 944 °C.

Chromium solubility limits were also measured in two low-level waste (LLW) glasses being developed by Pacific Northwest Laboratory (Li 1995). For LLW glasses L6-5412 and L4-9012, the solubility limits were found to be 0.48 and 1.04 wt% Cr_2O_3 , respectively, at 1,350 °C. The L6-5412 and L4-9012 LLW glasses are nearly identical in composition (56.78 wt% SiO_2 , 12 wt% Al_2O_3 , 0 wt% Fe_2O_3 , and 20 wt% Na_2O), except that L6-5412 contains 5 wt% B_2O_3 and L4-9012 9 wt% B_2O_3 . For aluminum silicate glasses such as HTB651, the solubility limit appears to be 0.8 wt% Cr_2O_3 at 1,350 °C (Li 1995). Spinel phases tend to crystallize from HTB651 when the temperature is reduced to 1,245 °C.

Chromium solubility limits have also been determined for high temperature borosilicate glasses that melt at temperatures from 1,314 to 1,328 °C (Vojtech et al. 1995). X-ray, SEM, and optical microscopy were used to identify Cr_2O_3 rich precipitates (eskolaites and spinel) in quenched and in thermally annealed samples with 1.1 wt% Cr_2O_3 and varying amounts of Fe_2O_3 , MnO , NiO , and ZrO_2 . Crystals were observed in four of five samples at their nominal melting temperature, and in all samples annealed for 24 hours 100 °C below the nominal melting temperature of the glass. The matrix glasses (uncrystallized portion of the sample) contained from 0.6 to 0.93 wt% Cr_2O_3 , with the highest level of matrix chrome being found in a glass with 2.5 wt% MnO_2 (compared to 0.7 wt% MnO in the other

samples). These results are important because they show that the solubility of Cr_2O_3 is only marginally increased by the temperature and composition of borosilicate and aluminum silicate glasses.

During melter studies, chromium rich spinel phases have commonly formed in glasses with 1.3 to 1.6 wt% Cr_2O_3 , but not in glasses with as little as 0.2 wt% Cr_2O_3 . Because of their small size, most spinel phases tend to remain in the melt. Settling can occur when these crystals form larger agglomerates. The settling characteristics of spinel phases were examined during a PSCM test. The melter feed contained 1.3 wt% Cr_2O_3 but about 5 percent of the Cr_2O_3 in the feed precipitated to the bottom of the melter (Perez and Nakaoka 1986). The product glass had a Cr_2O_3 concentration of 1.18 wt% while the bottom 2.5 cm of glass in the melter was found to contain 18.3 wt% Cr_2O_3 with some agglomerates as large as 80 μ . In the bottom section of the melter where the precipitates had settled, the glass matrix material contained 0.52 wt% Cr_2O_3 . This matrix concentration was found to be consistent with the Cr_2O_3 solubility limit of the HW39 glass used in the melter test.

One of the test melters at SRS also was affected by the accumulation of spinel phase. Spinel phases may have formed during low-temperature idling of the melter or perhaps because of corrosion of the Monofrax K-3 refractory materials in the melter. Such precipitates are known to have only a very minor effect on the durability of the glass (Bibler 1983, Bickford and Jantzen 1986, Zhu et al. 1986). Because spinels have little effect on properties of the glass, a Cr_2O_3 limit is needed mainly to avoid operability problems in the melter and more specifically to minimize the possible precipitation and accumulation of spinel. For the LTM and HTM melters, a Cr_2O_3 limit of 0.5 wt% in glass appears to be reasonable for flat-bottom melters where the accumulation of solids should be minimized. The limit at DWPF is 0.3 wt% Cr_2O_3 , that is consistent with the estimated composition envelop for the Phase I HLW glasses. This limit could be increased if reliable methods are developed to remove solids from the melter. Potentially viable methods include agitation or use of bottom-drain systems where the bottom is sloped to aid in removal of sludge.

5.7.2 Phosphorus Solubility

Phosphorus, or P_2O_5 , is relatively insoluble in borosilicate glasses at lower temperature. When such glasses are cooled, phosphorus tends to separate as a molten salt or particulate species (Volf 1984). Even small amounts of P_2O_5 (<0.5 percent) are apparently immiscible in silicate melts on the molecular scale. It is the coalescence of the immiscible phases into droplets that can create potential problems. Such droplets have been observed to form a "scum" on the surface of the glass melt that reduces throughput capacity of the melter. During the PSCM-19 melter test with WVDP 182 glass, an immiscible surface scum of calcium phosphate was formed (Bunnell 1988). This scum was absorbed through cold cap and coated the residual silica particles added as glass-forming material to the melt. It was found that this scum significantly reduced the melt-rate capacity of the melter. The WVDP 182 glass contained 3.0 wt% CaO, 3.2 wt% rare earths, and 2.7 wt% P_2O_5 . In a parallel test with WVDP 183 glass, a viscous surface scum formed that was found to be enriched in

rare earth phosphates, once again affecting the melting rate of the melter. The WVDP 183 glass contained 0.7 wt% CaO, 2.0 wt% rare earths, and 2.6 wt% P_2O_5 (Bunnel 1988).

Scoping studies have indicated that phase separation was caused by the accumulation of sulfate (gall) in WVDP 182 glass, inducing the separation of phosphate and other components from the glass. In crucible studies using oxides and carbonates, the WVDP 182 composition showed phosphate rich segregation after a gall layer formed from more than 1.5 wt% sulfate in the glass. This gall layer contained sodium, phosphorus, calcium, lanthanum, potassium and sulfur. Based on this evidence, it is assumed that P_2O_5 , CaO, Li_2O , and rare earths accumulate in the gall layer beneath the cold cap. After the sulfate volatilizes, a concentrated layer of crystalline calcium rare earth phosphates or other phosphates may be left on the surface of the melt.

This phenomenon is apparently reproducible, judging from the results of recent crucible tests at PNNL where phosphates interacted with rare earths to form an optically transparent secondary phase in quenched and CCC glasses. Such tests have indicated that the secondary phase consists of $Na_3RE(PO_4)_2$, Li_3PO_4 , or $NdPO_4$, where RE represents neodymium, cerium, or lanthanum. The sum of the rare earth components varied from 7.3 to 7.8 wt% and P_2O_5 from 3.0 to 9.0 wt% in these samples. The threshold for $NaRE(PO_4)_2$ precipitation in the CCC samples appears to be about 3.0 wt% P_2O_5 while the lower limit for Li_3PO_4 precipitation is about 3.0 wt% lithium and 3.0 wt% P_2O_5 . However, in one test, Li_3PO_4 crystals were observed by x-ray diffraction in a glass that contained 6.84 wt% Li_2O and only 1.35 wt% P_2O_5 . In CCC glasses, the durability was reduced by the precipitation of Al_3PO_4 from glasses with more than 5.5 wt% P_2O_5 and 8.3 wt% Al_2O_3 (Li 1995).

Glasses containing 4.26 wt% P_2O_5 and 0.25 wt% CaO were successfully processed in PNNL melters during PSCM-9. Aluminosilicate glasses with 20 wt% alumina, high-alkaline earth, and low-alkali metal have incorporated as much as 9.4 wt% P_2O_5 without salt separation and with good durability characteristics (Merrill and Janke 1993). High-alumina glasses are apparently capable of absorbing higher levels of phosphate. Li_3PO_4 phase separation was detected in SRS's SRL 165 borosilicate glass when more than 4.0 wt% PO_4 (3.0 wt% P_2O_5) was added to this composition (Jantzen 1986). Recently, PNNL formulated three borosilicate glasses (ECA 451, 452, and 501) based on an all-tank waste composition for the French low-temperature, induction-heated melter. These glasses contained about 1.0 wt% CaO, 6.6 wt% rare earths, and up to 2.56 wt% P_2O_5 but did not exhibit any evidence of phase separation or precipitation of phosphate crystals.

In summary, it appears that in some borosilicate glasses, phosphorus tends to form precipitates with calcium, rare earths, and lithium. Molten salt or scum layers may form from low-melting-temperature phosphates and interfere with the operation and melting rate of the melter. A limit of 3.0 wt% P_2O_5 should be acceptable because the Phase I glasses are projected to have less than 1.0 wt% P_2O_5 and 0.85 wt% RE, and should not be susceptible phosphate induced scum or phase separation in the melter. This limit is also consistent with the DWPF phosphate limit of 3.0 wt% PO_4 (2.24 wt% P_2O_5) (Hrma and Piepel 1994), recent test results just described, and limits derived from the WVDP 182 and 183 glasses.

5.7.3 Chloride and Fluoride Solubility

Halides are moderately soluble in silicate glasses at high temperature but separate as droplets or crystals (of NaCl, NaF or CaF₂) at lower temperatures producing optical opacity in the glass due to oversaturation. All halides have high-vapor pressures and may volatilize to a certain extent from the melt. Most commercial glasses contain about 0.6 wt% fluoride, at most, but the addition of alumina or boric oxide increases the solubility of fluoride to such an extent that about 7 percent of the oxygen can be replaced with fluoride in high-alumina glasses (Volf 1984). Chlorides are only sparingly soluble in glass with a maximum solubility of about 2.3 wt% as NaCl at 1,400 °C in sodium silicate glass and 1.5 wt% as KCl in lead silicate glass. In high-silica glasses, such as Pyrex¹ glass, the solubility limit for chloride is about 0.1 wt%.

The presence of sodium and boric oxides tends to increase the solubility of chloride in borosilicate glass. Pacific Northwest National Laboratory recently developed two glasses for the LLW program, L6-5412 with 5 wt% B₂O₃ and L4-9012 with 9 wt% B₂O₃. The chlorine solubility was found to be 0.57 wt% in L6-5412, and 0.49 wt% in L4-9012 at 1,350 °C, while fluorine solubility limits were 0.92 and 1.18 wt%, respectively. These limits did not vary significantly as a function of temperature over the range from 1,300 to 1,400 °C. The volatility of NaCl increases from 804 to 1,300 °C, with considerable loss from the melt at 1,300 °C. Fluorides generally volatilize as SiF₂ and BF₃ because the fluoride losses from borosilicate glasses are greater than those from alkali-lime-silica glasses. Also, chlorides and fluorides have not been linked to the corrosion of refractory materials in the melter. Apparently, refractory corrosion as such can be substantially suppressed with increased alumina content in the glass (Volf 1984).

Scoping tests were conducted to assess the effects of adding up to 5 wt% fluoride, 5 wt% BaO, 5 wt% MnO₂, 2 wt% MoO₃, and 2 wt% NiO to HW39 borosilicate glass (Larson 1989). During the performance of these tests, only 60 to 80 percent of the fluoride was retained in the sample glass. Except for fluoride, none of the other components had any significant effect on viscosity, electrical conductivity, phase behavior, or durability properties of HW39 glass. At 5 wt% fluoride, CaF₂ phase separation was observed in the glass. This condition is not acceptable because of potential durability problems and the possible impact of CaF₂ on melter operations. Glasses with 3.0 and 4.0 wt% fluoride were optically darker although regions of gross immiscibility were not found. Based on the increased release of sodium, boron, and silicon at 3.0 wt% fluoride during the MCC-3 durability test, an upper limit of 1.73 wt% fluoride was established for the HWVP reference glass.

Melter tests have shown that chloride losses can be high (typically about 25 to 50 percent of the feed chloride is lost by evaporation). About 83 wt% of submicron particles emitted during the PSCM-4 melter test consisted of NaCl. Offgas line deposits formed during melter idling contained 20-90 wt% NaCl and as much as 30 wt% SO₄. The PSCM-4 melter runs were performed under a variety of conditions (feed rate, plenum temperature,

¹Pyrex is a trademark of Corning Glass Works.

offgas flowrate, etc.), with feeds of different composition. Offgas deposits containing alkali borates, chlorides, fluorides, chromates, and sulfates were also produced by the Scale Glass Melter (SGM) at Savannah River (Jantzen 1991). Sodium and potassium chlorides, sulfates and borates were found on the interior canister walls, neck and shoulder above the melt line of canisters filled during campaign 10 of the SGM (Jantzen 1992).

During the PSCM-22 and PSCM-23 melter tests, that were conducted under similar conditions, variable amounts of chloride and fluoride were found in the product glass samples (Table 5-9) (Nakaoka 1985, Goles and Anderson 1986, Goles 1989, Goles and Nakaoka 1990).

These values are well below the solubility limits for chloride and fluoride in the literature and limits that were developed from crucible tests of HW39 reference glass. However, these values are useful because they demonstrate that chloride and fluoride are quite volatile, particularly in the presence of cesium and sodium oxides in the melt. The volatility of cesium is directly related to the concentration of chloride in the melt (Goles and Anderson 1986). Pacific Northwest National Laboratory has previously tested melter feeds with 0.3 wt% chloride (SRS), 0.35 wt% chloride (WVDP), and 0.08 wt% chloride (HWVP) (Anantatmula et al. 1991). Although halides can form molten salts that accumulate on the glass surface, only minimal quantities of such salts have ever been observed. There is no generally applicable solubility limit for chloride, although a solubility limit of less than 1.0 wt% NaCl is recommended for DWPF (Hrma and Piepel 1994). A limit of 0.5 wt% Cl is recommended based on experimental results from the LLW program.

Table 5-11. Chloride and Fluoride Glass Compositions and Percent Retained During Pilot-Scale Ceramic Melter-22 and Pilot-Scale Ceramic Melter-23 Melter Tests.

Melter test	Chloride, p/M	Chloride, percent retained	Fluoride, p/M	Fluoride, percent retained
PSCM-22	148	19	550	27
PSCM-23	948	85	1,998	90

PSCM = Pilot-scale ceramic melter.

5.7.4 Sulfate Solubility

Sulfate is relatively insoluble in borosilicate glass. Molten salt separation, that reduces (or increases) melter throughput capacity, could occur. Poor durability characteristics may be found in glasses with high-sulfate content, particularly if they tend to form a gall or molten sodium sulfate layer on the glass surface. Such layers have been found in some of the test melts with Fernald wastes (Pegg et al. 1994). A molten salt layer could readily

incorporate radionuclides such as cesium, and by doing so, increase the volatility of cesium and possible entrainment losses from the melter. Segregated melts (with liquid sodium sulfate) may extract components that would otherwise remain in the molten glass, such as phosphates, chromates, molybdates and halides. Molten salts also tend to dissolve the refractory and accelerate refractory corrosion processes at the melt line.

Sulfate solubility in glass is sensitive to the oxidation state and basicity of the glass melt. In a study of sulfate in borosilicate glass, the $\text{Fe}^{+2}/\text{Fe}^{+3}$ ratio was systematically varied from 0 to 2.5 for glasses in which 0.5 and 1.0 wt% sulfate were added to NCAW at 25 percent WOL (Larson 1989, Bates et al. 1985). The sulfate solubility did not change for the sample with 0.5 wt% sulfate over the REDOX range of interest (0.0 to 0.3). In contrast, all the glass samples with 1.0 wt% sulfate had a yellow or white surface layer on the glass indicating a limited amount of phase separation at 1.0 wt% sulfate. The sulfate solubility limit was determined to be 0.9 wt% for these samples over the REDOX range in the melter (0.0 to 0.4). Sulfate solubilities decreased about 50 percent as the REDOX ratio increased from 0.5 to 1.0 (reflecting the development of more reducing conditions in the melter). Pacific Northwest National Laboratory recently developed two glasses for the low level waste (LLW) program, L6-5412 with 4 wt% CaO , and L4-9012 with 0 wt% CaO . The sulfate solubility limit was found to be 0.75 wt% in L6-5412, and 0.47 wt% in L4-9012 at 1,350 °C, reflecting the favorable impact of calcium on the sulfate solubility limits in borosilicate glass. The HWVP sulfate limit is 0.5 wt% as SO_3 while the DWPF sulfate limit is 0.4 wt%. Both limits are based on the desire to limit sodium sulfate phase separation from the glass.

5.7.5 Bismuth and Uranium Solubility

Currently, there are no solubility limits for bismuth oxide (Bi_2O_3) that can be applied to Phase I glasses. Bismuth oxide is a conditional glass former like Al_2O_3 . Bismuth has properties in glass most similar to lead due to similarities in electronic configuration and molecular weight (Volf 1984). Bismuth glasses are reported to strongly attack refractory materials. Under reducing conditions, bismuth glasses tend to form a metallic mirror on the surface of the glass. This is not expected to occur in the HLW melter because bismuth is relatively dilute in the HLW. Glass viscosities could be reduced and durability improved with the addition of Bi_2O_3 . Because of the low bismuth content of Phase I wastes, this component is not expected to effect the behavior of the glass.

Samples of first and second cycle bismuth phosphate waste simulant were vitrified in 1978 to determine the characteristics of this glass (Kupfer 1978). These glasses contained as much as 13 wt% Bi_2O_3 at 40 percent waste oxide loading. Glasses were found to be clear dark green-brown and durable when tested by the soxhlet leach method.

Pacific Northwest Laboratory conducted a single component study of bismuth by varying the concentration of bismuth from 1 to 15 percent in high-temperature

nonborosilicate glass. The results confirm those of Kupfer (1978). Bi_2O_3 decreases viscosity, increases durability, and slightly decreases T_L for spinel forming glasses.

Uranium is usually present in the tetravalent and hexavalent state in glass. The hexavalent form is the more stable oxide and is readily soluble in glass while the tetravalent form has only limited solubility (Schreiber and Balazs 1982, Schreiber et al. 1983). The presence of fluoride in uranium-containing glasses is undesirable because uranium fluorides (UF_6) are quite volatile. Alkali-borate and phosphate glasses may absorb as much as 60 wt% UO_2 under oxidizing conditions. In lithium-barium-silicate glasses, UO_2 improves the resistance to devitrification and permits the formulation of glasses that would otherwise be susceptible to crystallization (Volf 1984).

In a single-component study, no evidence of uranium phase separation was observed over the range of uranium compositions tested (0 to 8 wt%). The normalized uranium release increased from 2.4 to 5.1 g/m^2 as the UO_3 concentration increased from 0.15 to 8 wt% in MCC-1 durability tests. Releases of silicon, boron, lithium, and sodium were 50 percent below EA glass durability standards over this range. Also, the amount of iron in chromite spinel actually decreased with an increase in UO_3 . In glasses with more than 2 wt% UO_3 , the crystalline phase was determined to be Cr_2O_3 . The viscosity and electrical conductivity properties of borosilicate glass were not affected by UO_2 concentrations of up to 8 wt%. Pacific Northwest Laboratory conducted a parametric study to investigate the effects of uranium on the properties of glass. Uranium was completely soluble to a level of 15 wt% in glass. UO_2 slightly decreases glass viscosity and had no measurable effect on glass durability. Uranium does not appear to be a limiting component in HLW glasses at this time.

5.7.6 Noble Metals Solubility

Precipitation of silver and noble metals from the glass is one of the most significant feed processability issues for the Phase I feeds. Noble metal oxides and alloys (RuO_2 , PdO , Rh_2O_3 , Ag_2O) are only slightly soluble in borosilicate glass. The solubility limit in silicate glass is approximately 0.05 wt% for rhodium and 0.03 wt% for palladium (Volf 1984). RuO_2 has an apparent solubility limit of 0.01 wt% in borosilicate glasses with less than 25 wt% Na_2O , but it has a limit of 0.25 wt% when Na_2O exceeds 25 wt% (Mukerji 1972). Some investigators have found that the solubility of noble metals is controlled by the REDOX state of the glass (Schreiber et al. 1991). Significant noble metal-rich deposits have accumulated to form an electrically conductive sludge layer at the bottom of U.S. melters and melters in Germany and Japan (Hutson 1994, Grunewald 1993a, Cooper et al. 1994). This sludge has caused short circuiting of lower electrodes in flat-bottom test melters and failure of at least one bottom-drain system. The conditions under which these deposits have formed have been extensively studied to characterize the nature and morphology of these deposits. Computer simulation studies also have been performed to predict expected melter lifetime based on the accumulation of noble metals when processing high-burnup nuclear wastes such as NCAW DST waste.

In 1976, Germany began developing a vitrification program for treating radioactive wastes. Most of the initial work was performed on wastes generated from reprocessing fuel at the Eurochemie plant in Mol, Belgium. The low enriched-waste concentrate had a high concentration of noble metals that settled to the floor of the PAMELA melter to form a conductive layer leading to processing difficulties in the melter (Elliott et al. 1994). During the first year of operation, PAMELA produced 47 m³ of vitrified waste. Melter feed rates decreased from 30 to 20 kg/h after the first year due to the accumulation of noble metals on the floor of the melter. This layer had an electrical conductivity three times higher than the bulk glass. Because the settled sludge layer was significantly more conductive than the bulk glass, the current field and electrical potential were affected, limiting processing capacity of the melter. Table 5-12 shows the noble metals concentration in the PAMELA feed and retention of noble metals in the melter.

Table 5-12. Noble Metals Feed Composition and Percent Retained in the PAMELA Melter.

Component	PAMELA feed, p/M	Percent retained in melter
Ruthenium	1,000 to 5,100	35.1
Rhodium	3,000	21.2
Palladium	1,400	-0.2

Assuming 25 percent WOL, the PAMELA melter glass retained 0.12 to 0.17 wt% noble metals.

A sample of the PAMELA melter bottom glass showed areas of ruthenium concentration 80 times higher than the concentration in the bulk glass. Approximately 215 m³ of highly enriched waste concentrate (with fewer noble metals) was processed through PAMELA during the last campaign. Altogether, four electrode failures were observed. Some believe they may have been caused by electrical shorting of the bottom electrodes while others attribute the failures to electrical asymmetry effects in the melter. In November 1986, the bottom drain failed with the suspected cause being the accumulation of noble metal deposits on the floor of the melter. Finally, in 1988, the PAMELA overflow system failed, leading to the shutdown and replacement of the melter.

Two melter test campaigns were conducted at SRS for the HWVP project using NCAW simulants with noble metals (Hutson 1994). Table 5-13 shows the approximate noble metals composition of the feed and the percent retained in the melter (Elliott et al. 1994).

Table 5-13. Noble Metals Feed Composition and Percent Retained in the Savannah River Site Test Melter (IDMS Melter).

Component	Feed composition (wt%)			Percent retained in melter	
	HWVP1	HWVP2	NCAW 1991	HWVP1	HWVP2
Ru_2O_3	0.17	0.05	0.11	13.1	12.7
Rh_2O_3	0.05	0.01	0.03	15.1	9.2
PdO	0.09	0.01	0.03	3.4	-8.1

DWPF = Defense Waste Processing Facility

HWVP = Hanford Waste Vitrification Plant

IDMS = Integrated DWPF Melter System

NCAW = Neutralized current acid waste.

At an assumed 28 wt% WOL, HWVP1 and HWVP2 glasses retained 0.077 and 0.017 wt% noble metals, respectively.

A small research scale (1/100-scale) melter test was performed in FY 1991 to assess the impact of noble metals on melter operation (Cooper et al. 1994). The nominal feed consisted of NCAW simulant with the appropriate concentration of noble metals. During one of the latter segments, the feed concentration of noble metals was doubled so the impact of feed concentration on settling behavior could be evaluated. Toward the end of the melter campaign, the resistance between electrodes decreased. Upon destructive examination of the melter, a layer of noble metals was found on the bottom of the melter and approximately one-third of one electrode was lost due to corrosion or localized overheating. The metal layer at the bottom also removed approximately 1.3 cm of the bottom refractory. This also has been observed in commercial melters where bottom refractories have been attacked by downward drilling of molten metals. Mass balance calculations indicate that approximately 5 percent of the noble metals present in the feed precipitated to the floor during segments of the test with nominal noble metals concentration (0.332 wt%, including Ag_2O), and 46 percent precipitated during segments with higher noble metals concentration (0.465 wt%) in the melter feed. The agglomeration and precipitation of noble metals occurred over an average residence time of only 5 hours in the melter. Glass samples taken from the bottom of the research-scale melter contained clusters of RuO_2 needles and metallic particles of Ru/Rh surrounded by layers of Pd/Ag/Te alloy and RuO_2 (Cooper et al. 1994).

An engineering-scale (1/10-scale) melter test was conducted at Kfk in FY 1992 to evaluate the behavior of noble metals during prolonged melter operation at the projected plant operating conditions (Grunewald 1993b). The melter feed consisted of NCAW simulant with the normal noble metals content. During the 49 days of testing with NCAW, about 35 percent of the ruthenium, palladium, and rhodium fed to the melter settled to the melter floor. Samples taken from the bottom of the melter at the end of the run had high concentrations of noble metals, 20 to 45 times those of the nominal glass. Near the end of the run, the electrical resistance between the lower set of electrodes decreased by 10 to

15 percent, consistent with the presence of a noble metals layer on the floor of the melter. In a related study, it was observed that the solubility of palladium increases with oxygen fugacity or concentration in the glass, but at constant oxygen fugacity the solubility of palladium increased only slightly with temperature.

During the most recent series of large-scale melter tests in Japan, a total of 40.9 m³ of simulated HLW was processed. Noble metal composition in the melter feed, retention rates, and average glass compositions are presented in Table 5-14.

Table 5-14. Noble Metals Feed Composition, Percent Retained, and Average Glass Composition in Large-Scale Japanese Melter Tests.

Component	Melter feed (wt %)	Percent retained	Percent glass composition
RuO ₂	2.25	0.08	0.87
PdO	1.06	0.09	0.51

In contrast to earlier melter campaigns, these results were obtained from a sloped-bottom melter with a bottom-drain valve. In general, the Japanese have observed that the accumulation rate of noble metals can be sharply reduced by increasing the slope of the melter floor and using a bottom-drain system. This observation is important because it suggests that the shape and design of the melter have a significant effect on the accumulation rate and acceptable solubility limit for noble metals.

The preliminary limit for HWVP noble metals is less than 0.25 wt% Ru₂O₃+PdO+Rh₂O₃ in glass (Kalia 1992a). This limit is well above solubility limits derived from other studies, that indicate flat-bottom melters may be susceptible to premature electrical failure at noble metals concentrations as low as 0.1 wt% in glass.

5.7.7 Titanium Solubility

Titanium oxides can adversely affect the homogeneity of glass by inducing precipitation of spinel and lithium and sodium aluminum silicates. Titanium oxide also reduces glass viscosity and enhances the solubility of actinides. As the TiO₂ concentration is raised from 0 to 10 wt%, the solubility of uranium in borosilicate glass increases from 15 to 30 wt% while the solubility of plutonium increases from 2 to 4 wt% (Plodinec 1979). However, the main concern is the impact of TiO₂ on devitrification of the glass. The primary crystalline phases formed after the addition of TiO₂ are alkali aluminosilicates.

Titanium oxide is commonly used as a nucleating agent in lithium and magnesium aluminosilicate glasses for the production of glass ceramics. Titanium oxide is effective for inducing the precipitation of alkali aluminosilicates because of its poor solubility in alumina rich glasses. About 3 to 4 wt% TiO_2 is normally used to nucleate glass ceramics in $\text{Li}_2\text{O}-\text{Al}_2\text{O}_3-\text{SiO}_2$ glasses, while 8 to 10 wt% TiO_2 is required for $\text{MgO}-\text{Al}_2\text{O}_3-\text{SiO}_2$ glasses. Formation of these phases could seriously affect the operation of the melter (Plodinec 1979, Plodinec and Wilds 1978). Rutile (TiO_2) crystallized in the PAMELA melter outlet channel and pour spout when glasses with 1.55 wt% TiO_2 , 20.68 wt% Al_2O_3 , 21.7 wt% B_2O_3 , and 3.1 wt% Li_2O were melted. The bottom drain plugged several times and rutile crystals formed in the glass that remained in the outlet channel for some time (Langowski 1994).

When aluminosilicates dissolve, the viscosity of glass in the vicinity of the crystals increases, retarding or slowing dissolution of the crystals. If this material is formed in the throat of a joule-heated melter, it could seriously affect the glass-pouring operation and overall melter performance. Even if this material passes through the melter, it is likely to be rich in radiocesium and highly leachable. In addition to other factors, the depletion of aluminum in the glass matrix due to the precipitation of lithium aluminum silicate may reduce the durability of certain glasses. An almost linear relationship exists between the extent of devitrification and the TiO_2 content in glass (Plodinec 1979). Because the rate and extent of devitrification depend directly on TiO_2 concentration, DWPF glasses are limited to a maximum concentration of 1.0 wt% TiO_2 . The DWPF glasses are currently projected to have 4 to 7 wt% Al_2O_3 , 3 to 5 wt% Li_2O , and 8 to 12 wt% Na_2O and B_2O_3 (WSRC 1992).

In contrast to recent experience with borosilicate glasses, no evidence of crystallinity was found in alumina-silicate glasses with 1.0 to 8.0 wt% TiO_2 (HTB651-7 through -12), after these glasses were quenched from 1,350 °C (Li 1995). The HTB651 series of glasses contains 16 to 17 wt% Na_2O and 8.4 to 12.0 wt% Al_2O_3 , but no Li_2O . Based the phase equilibrium diagram for the $\text{Na}_2\text{O}-\text{Al}_2\text{O}_3-\text{SiO}_2$ system, nepheline would have precipitated from such glasses with 19 to 29 wt% Na_2O , 29 to 30 wt% Al_2O_3 , and 45 to 55 wt% SiO_2 at 1,350 °C. The acceptable concentration limits for TiO_2 are inevitably linked to the concentration of other components in the glass, especially Al_2O_3 , Li_2O , B_2O_3 and Na_2O , as these components affect the precipitation of alkali aluminosilicates. Based on presently available information, a TiO_2 limit of 3 wt% TiO_2 in low (2 to 3 wt%) Li_2O borosilicate glasses, and 5 wt% TiO_2 in low (2 to 3 wt%) Li_2O aluminosilicate glasses is believed to be acceptable, pending the completion of more definitive studies to determine the precise glass composition limits for TiO_2 .

5.7.8 Minor Component Solubility Limits

Table 5-15 summarizes the minor component solubility limits recommended in this study and compares these limits to those developed for HWVP and DWPF.

Table 5-15. Minor Component Glass Solubility Limits for High-Level Waste.

Component	Current study (wt%)	HWVP (wt%)	DWPF (wt%)
Cr ₂ O ₃	< 0.5 ^a	< 0.5	< 0.3
P ₂ O ₅	< 3.0 ^b	< 1.0	< 3.0 (as PO ₄)
SO ₃	< 0.5	< 0.5	< 0.4 (as SO ₄)
Cl	~ 0.5	no limit	< 1.0 (as NaCl)
F	< 1.7	< 1.7	< 1.0 (as NaF)
Bi ₂ O ₃	< 13.0	no limit	no limit
UO ₃	< 15.0	no limit	no limit
Ru ₂ O ₃ + PdO + Rh ₂ O ₃ + Ag ₂ O	< 0.1-0.2 ^c	< 0.25	no limit
TiO ₂	< 3.0 (Borosilicate), < 5.0 (aluminosilicate)	no limit	< 1.0
Cu	no limit	no limit	< 0.3

DWPF = Defense Waste Processing Facility

HWVP = Hanford Waste Vitrification Plant

^aThe Cr₂O₃ solubility limit may increase to 1.0 wt% for low-iron borosilicate and nonborosilicate glasses depending on the results of glass and melter feature development testing.

^bLimit for P₂O₅ based on commensurate limits of < 3.0 wt% Li₂O, < 1.0 wt% CaO, and < 1.5 wt% rare earths (cerium, lanthanum, and neodymium).

^cFlat-bottom melter with no bottom drain.

5.8 COMPARISON OF SIMULANTS TO RADIOACTIVE PHASE I WASTES

Most studies have relied on simulants to represent the properties of radioactive waste. Due to the relative costs involved, cold simulants have normally been used for the feed chemistry, HLW glass formulation and melter development programs at the Hanford Site, SRS, WVDP, and abroad. Laboratory-scale tests with radioactive waste are therefore needed to establish a credible link between simulant based studies and the behavior of actual tank wastes during feed preparation and vitrification of such wastes. This comparison not only helps to validate simulant tests, but also provides a basis for validating the glass property models used to formulate HLW glasses of acceptable composition.

Three core samples of radioactive waste were recovered from Phase I Tanks 241-AZ-101 and 241-AZ-102 in 1989. Small portions were used in laboratory-scale feed preparation and vitrification tests to produce representative samples of radioactive glass. The next section will briefly describe the nature of this testing program and will compare the properties of the radioactive samples to simulants and glass model predictions for this waste (Morrey 1995).

The solids from each core were taken through a prescribed pretreatment process for Phase I wastes, using a water wash/settle/decant sequence, with ferric-nitrate flocculent, and two water washes (3 volumes deionized water to each volume of sludge). These samples were then analyzed to determine the composition of the washed solids. Afterwards, the washed solids were adjusted to an average concentration of 125 g waste oxide/L and treated with formic acid to adjust feed rheology and to chemically reduce REDOX-sensitive species (Fe, Mn, etc.) in the waste. Frit was then added to simulate the last step in the feed preparation process. The formatted slurry and frit/slurry samples were thoroughly characterized as to their chemical, radiochemical, physical and rheological properties. Frit/slurry mixtures were dried and melted at 1,150 °C in crucibles and resulting glasses were analyzed to determine their chemical and radiochemical composition, durability (PCT and MCC-1 tests), crystallinity, REDOX state and density.

5.8.1 Slurry Properties

The main components in the waste are Fe, Al, and Na, present in the form of OH^- , CO_3^{2-} , NO_3^- , and NO_2^- .

Slurry simulants were prepared using PNNL's procedure (WTC-006-36). In this procedure, major components such as Al, Fe, Mn, Ni, and Zr are precipitated with NaOH from nitrate solutions and washed to remove sodium and nitrate. Insoluble minor components (Ag, Cd, Ce, Cr, La, etc.) are co-precipitated, washed and blended with major components. Soluble and slightly soluble components (B, Ba, Ca, Cs, Cu, Na, etc.) are added as oxides, hydroxides, fluorides and sulfates, while Na is added in multiple forms to match carbonate and anion fractions in the waste. Simulant compositions were designed to match the composition of the radioactive samples.

During the formatting process, CO_3^{2-} , NO_3^- , and NO_2^- are converted into CO_2 , NO_x , and other nitrogen compounds (N_2 , N_2O , NH_3), while the concentration of these anions in the sample is reduced. Slurry chemistry and offgas generation reactions were found to be similar for core samples and simulants, with any differences being explained by testing conditions and slurry chemical composition effects. The only radionuclides affected by the washing step were ^{137}Cs and ^{125}Sb , while ^{129}I was the only radionuclide that may have been lost during the formatting step. Settling behavior of the simulants did not match the behavior of the core samples in that the core samples settled much more quickly and achieved a much higher solids density than the simulants. Previous studies have shown that when the core samples are allowed to dry, the settling and rheological properties of the sample can be irreversibly changed. However, the particle size distribution of the core samples did not change during the feed preparation process. Most particles were less than 5 microns in diameter, with a significant fraction less than 1 micron.

The rheology properties of the radioactive waste were also different from those of the simulant. Formic acid treated wastes had a lower yield stress and lower apparent viscosity than the formatted simulant. The rheology characteristics of the waste can best be described

as pseudoplastic yield behavior with slight shear-thinning and some hysteresis. The rheology properties of the wastes and simulants are compared in Table 5-16. At Savannah River, yield stresses were reduced and settling rates increased for radioactive (potassium tetraphenylborate) slurries, compared to nonradioactive simulants, because of radiation induced de-agglomeration of the sludge (Morrey 1995). These effects could also be responsible for the differences between the Phase I wastes and simulants.

Table 5-16. Comparison of Radioactive Sample and Simulant Rheology Properties.

	Radioactive Samples			Simulants		
	Yield Stress, Pa	Apparent Viscosity @ 50 s ⁻¹ , cP	Total oxides, g Wo/L	Yield Stress, Pa	Apparent Viscosity @ 50 s ⁻¹ , cP	Total oxides, g Wo/L
Formatted slurry	0.85 to 0.23	3.8 to 7.2	101 to 147	1.2	32	147 to 161
Frit/formatted slurry	1.4 to 10.3	38 to 260	479 to 600	2.2 to 12.4	58 to 365	438 to 573

5.8.2 Glass Properties

Based on standard KOH/Na₂O₂ fusion and ICP/AES analysis procedures, the measured analytes were generally found to be within 10 percent of the target values predicted from the washed solids composition, frit compositions and estimated waste loadings for the glasses. The REDOX ratio (Fe²/Fe³) varied from 0.026 to 0.085 for the radioactive glasses, compared to 0.005 for the simulant. The radioactive glasses were analyzed by X-ray diffraction to determine the composition of the crystalline phases. Since the radioactive glasses were formulated to a liquidus temperature of less than 900 °C, very little if any crystallinity was found in the quenched samples (less than 1 vol%). In one sample from Tank 241-AZ-101, ruthenium oxide was tentatively identified as a solid phase of minor significance. The density of the radioactive glasses ranged from 2.56 g/cc for Tank 241-AZ-101 core 1, to 2.67 g/cc for Tank 241-AZ-101 core 2, and 2.54 g/cc for Tank 241-AZ-102 at room temperature, densities that are consistent with the simulant glasses for these wastes.

Each of the radioactive and simulant glasses proved to be a highly durable waste form, at least 15 times more durable than the Savannah River EA glass as measured by PCT (Table 5-9). The durability results and model predictions are summarized in Tables 5-17, 5-18, and 5-19. Seven-day B releases for the radioactive glasses ranged from 0.13 to 0.22 g/m², compared to simulant boron releases of 0.2 to 0.34 g/m². The CVS second-order mixture model #3 for durability, described in Section 5.6 and Table 5-10, tends to over-predict releases in high-durability glasses and under-predict releases in low-durability glasses.

Table 5-17. Durability Results for 241-AZ-101, Core 1 Glass.
(Mean Normalized Release, g/m²)

Component	Radioactive glass	Simulant glass	CVS model (2 nd order, #3)
B	0.130	0.200	0.4801
Li	0.226	0.245	0.5008
Na	0.144	0.199	0.3589
Si	0.077	0.129	

Table 5-18. Durability Results for 241-AZ-101, Core 2 Glass.
(Mean Normalized Release, g/m²)

Component	Radioactive glass	Simulant glass	CVS model (2 nd order, #3)
B	0.223	0.338	0.4364
Li	0.293	0.369	0.4212
Na	0.105	0.248	0.2948
Si	0.141	0.184	

Table 5-19. Durability Results for 241-AZ-102, Core 1 Glass.
(Mean Normalized Release, g/m²)

Component	Radioactive glass	Simulant glass	CVS model (2 nd order, #3)
B	0.211	0.271	0.6775
Li	0.301	0.351	0.6338
Na	0.049	0.130	0.3971
Si	0.158	0.163	

Radiation has a significant effect on glass corrosion in aqueous leaching tests. Radiation effects can be mitigated by using Ar in the leach containers to minimize radiolytic nitric acid production. The dominant corrosion mechanism appears to be network hydrolysis that occurs under higher pH conditions. Seven day PCT and 28 day MCC-1 radionuclide releases were measured. As with previous studies, normalized releases of Am ranged from 0.1 to 6 percent of the B releases. Also consistent with previous studies, Tc, U, Np, and Cs were generally more soluble than Am (> 10 percent of the B-normalized release). However, Pu releases were similar to those of B, while Sr was relatively soluble in MCC-1 and insoluble in the PCT tests. The durability results are based on triplicate samples and standard reference glasses, with tests performed in-cell and out-of-cell in fused-silica-lined stainless steel and teflon containers, respectively.

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6.0 PROCESSABILITY ASSESSMENT FOR PHASE I FEEDS

In this section, optimum WOL and glass composition estimates are developed for representative Phase I feeds as a function of composition for the LTM. These estimates are be used as a basis for predicting the volume of glass produced from the Phase I wastes. These data are needed for development of the *Phase I High-Level Waste Pretreatment and Feed Staging Plan* (Manuel et al. 1996) and to assess the need for caustic washing and for specific blending strategies to minimize the potential volume of Phase I HLW glasses. These glass composition estimates, provided by PNNL, are based on current glass property data correlations (models) for borosilicate glass, especially the first and second order models that were developed during the composition variability study (CVS) at the Hanford Site (Hrma 1994). Because of the importance of glass volume, sensitivity studies were performed to evaluate effects of feed composition and more aggressive caustic washing processes for these wastes.

The composition limits of the CVS study (or composition range of glasses in the CVS) have sometimes been used to estimate the composition limits for HLW glasses, without assurance that these glasses also meet property constraints for the LTM. A much more reliable basis for evaluating glass composition is to use empirical glass property data and property models for borosilicate glass. These models can be used to estimate the most important properties of the glass based on the combined effect of all of the major components in the glass. Generally, good empirical models for viscosity, electrical conductivity and PCT releases are available, but comparable models for liquidus temperature are still being developed. Liquidus temperature is likely to be the property that limits waste loading for Phase I privatization feeds. It was, therefore, necessary to obtain additional liquidus temperature data for the Phase I glasses.

The balance of process issues also will be addressed in this section, including an assessment of volatile components that may affect the performance of the melter offgas system. Other operability issues include corrosion of offgas and feed preparation system components, nitrogen oxide emissions and criticality control.

6.1 WASTE COMPOSITIONS FOR GLASS FORMULATION DEVELOPMENT

The wastes compositions of interest include water washed (WW) and caustic washed (CW) sludges from Tanks 241-AZ-101, 241-AZ-102, 241-AY-102, and 241-C-106. For glass formulation purposes, several blends were also considered, including a blend of Tanks 241-AZ-101 and 241-AZ-102 (AZ blend), and a nominal blend (consisting of 80 percent of Tank 241-AZ-101, 80 percent of Tank 241-AZ-102, 32 percent of Tank 241-C-106, and 15 percent of Tank 241-AY-102 wastes). The nominal blend is assumed to have 32 percent of the waste in Tank 241-C-106, while the Phase I RFP nominal blend contains 37 percent of the Tank 241-C-106 waste. This difference as to the assumed retrieval fraction for Tank 241-C-106 has an insignificant effect on composition of the glass. The Phase I waste

compositions were derived from the Tank Characterization Reports for Tanks 241-AZ-101, 241-AZ-102, and 241-AY-102, except for the sludge and aluminum inventories in Tank 241-AZ-101. Composition estimates for Tank 241-C-106 are based on the Data Transmittal Package for Tank 241-C-106 (Weiss 1986). Because of possible uncertainties in the Tank 241-C-106 aluminum inventory, two separate estimates are provided for this waste (241-C-106A from Weiss and 241-C-106B based on Hara) (Weiss 1986, Hara 1990). Water and caustic washed sludge compositions were developed from core sample washing studies, analytical results for Tank 241-C-106 waste and from caustic washing results for Tank 241-C-103 waste (as the best available surrogate for Tank 241-C-106 waste) (Peterson 1989, Gray 1993a, Gray 1993b, Weiss 1986, and Rapko 1995).

Feed compositions reported in Manuel et al. (1996) are slightly different from those used in this study for glass formulation purposes. These differences do not materially affect the feed processability evaluation and are bounded by sensitivity studies in this report.

To bracket the expected range of glass compositions, PNNL has experimentally developed optimum glass compositions for the water and caustic washed nominal blend, and water washed 241-C-106A, 241-C-106B, and AZ blend. Water washing removes water soluble species and caustic washing, performed in addition to water washing, decreases the aluminum, chromium and phosphorus in the waste. The waste compositions used for glass formulation are listed in Table 6-1.

Table 6-1. Waste and Blend Compositions in Mass Fractions on Nonvolatile Oxide Basis.
Only components of ≥ 0.1 wt% are included in this table

Oxide	Nominal-W	Nominal-C	C106A-W	C106B-W	AZ-W
Ag ₂ O	0.0015	0.0016	0.0012	0.0037	0.0013
Al ₂ O ₃	0.1684	0.0960	0.1618	0.3555	0.1739
As ₂ O ₃	0.0009	0.0009	0.0000	0.0008	0.0017
B ₂ O ₃	0.0014	0.0015	0.0001	0.0014	0.0022
BaO	0.0062	0.0068	0.0114	0.0006	0.0015
BeO	0.0014	0.0015	0.0028	0.0000	0.0001
CaO	0.0212	0.0232	0.0348	0.0043	0.0081
CdO	0.0142	0.0156	0.0009	0.0000	0.0268
CeO ₂	0.0014	0.0015	0.0000	0.0000	0.0025
Cr ₂ O ₃	0.0045	0.0044	0.0030	0.0042	0.0057
CoO	0.0019	0.0021	0.0000	0.0000	0.0037
CuO	0.0007	0.0008	0.0003	0.0006	0.0010
F-	0.0012	0.0013	0.0000	0.0000	0.0019
Fe ₂ O ₃	0.2662	0.2918	0.1559	0.1929	0.3656
K ₂ O	0.0060	0.0066	0.0033	0.0000	0.0084
La ₂ O ₃	0.0057	0.0062	0.0000	0.0000	0.0107
MgO	0.0126	0.0138	0.0228	0.0011	0.0029
MnO	0.0069	0.0076	0.0050	0.0095	0.0082
Na ₂ O	0.2190	0.2400	0.2515	0.2916	0.1915
Nd ₂ O ₃	0.0038	0.0041	0.0000	0.0000	0.0071
NiO	0.0110	0.0121	0.0025	0.0000	0.0188
P ₂ O ₅	0.0097	0.0036	0.0119	0.0232	0.0072
PbO	0.0035	0.0039	0.0053	0.0043	0.0020
Rh ₂ O ₃	0.0006	0.0006	0.0000	0.0000	0.0011
RuO ₂	0.0007	0.0008	0.0000	0.0000	0.0013
Sb ₂ O ₃	0.0015	0.0016	0.0000	0.0057	0.0028
SeO ₂	0.0019	0.0021	0.0000	0.0030	0.0037
SiO ₂	0.1626	0.1782	0.3182	0.0911	0.0232
SrO	0.0006	0.0007	0.0003	0.0002	0.0010
SO ₃	0.0040	0.0044	0.0000	0.0000	0.0077
TeO ₂	0.0014	0.0015	0.0000	0.0000	0.0027
ThO ₂	0.0006	0.0007	0.0000	0.0000	0.0012
TiO ₂	0.0006	0.0007	0.0000	0.0009	0.0011
Tl ₂ O	0.0032	0.0035	0.0000	0.0020	0.0062
U ₃ O ₈	0.0135	0.0148	0.0010	0.0018	0.0242
ZrO ₂	0.0384	0.0421	0.0058	0.0013	0.0690
M _w (Mg)	396	358	486	310	282

M_w denotes waste mass in megagrams (metric tons).

6.2 GLASS FORMULATION

This section summarizes work to develop glass formulations and glass property estimates for Phase I glasses. Phase I glasses developed by PNNL were formulated for a low-temperature, joule-heated melter (LTM) operating at 1,150 °C. Glass properties of most importance are viscosity, electrical conductivity, liquidus temperature, and leach resistance or durability properties of the glass. The Phase I glasses are formulated for a viscosity and electrical conductivity of 5.0 Pa-s and 0.18 to 0.50 S/cm at 1,150 °C, respectively, a liquidus temperature of less than 1,050 °C (at least 100 °C below the nominal operating temperature of the melter), and boron and sodium releases, as determined by the PCT test, of less than 2.0 g/m². Liquidus temperature is especially important because this is the highest temperature where the melt is in equilibrium with the primary crystalline phase (at lower temperatures, possible sludge forming conditions could develop in the melter). Liquidus temperature appears to be the limiting constraint for most Phase I glasses. The wastes in Table 6-1 contain a significant inventory of refractory oxides (Al₂O₃, Fe₂O₃ and ZrO₂) that promote crystallization. As previous studies have shown, the optimum glasses developed from these wastes are likely to be limited by liquidus temperature criteria. High Al₂O₃ concentrations can also cause the crystallization of nepheline (NaAlSi₃O₈) in the canister, which adversely affects the durability of the glass (Kim et al. 1995). Borosilicate glasses typically contain at least 5 wt% B₂O₃, as a defined borosilicate waste form. The optimum glass is one that allows the maximum waste oxide loading in the glass subject to the following conditions:

- SiO₂: Glasses may become structurally unstable if the SiO₂ concentration is reduced to less than approximately 37 wt%. This limit is recommended for glass formulation purposes because little information is available on component effects at lower SiO₂ concentration. The optimum SiO₂ concentration also depends on the composition of other components in the glass. The CVS range for SiO₂ is 42 to 57 wt% (Table 5-3).
- B₂O₃: Low-temperature glasses should have more than 5 wt% B₂O₃ to meet the minimum standards for borosilicate glass, as defined in the WASRD glossary. The CVS range for B₂O₃ is 5 to 20 wt% (Table 5-3).
- Al₂O₃: The CVS range for Al₂O₃ is presently 0 to 17 wt% (Table 5-3). High Al₂O₃ concentrations increase the potential for nepheline precipitation in the canister (above 14 to 17 wt% Al₂O₃) and lithium aluminum silicate precipitation in the melter.
- Fe₂O₃: The CVS range for Fe₂O₃ is 0.5 to 15 wt% (Table 5-3). High Fe₂O₃ in combination with Cr₂O₃, MnO, and NiO increases the risk of spinel precipitation in the melter, as measured by the liquidus temperature of the glass.

ZrO ₂ :	For ZrO ₂ , the CVS range is currently 0 to 13 wt%, although zirconium containing phases often precipitate from borosilicate glasses with more than 7 to 10 wt% ZrO ₂ .
Na ₂ O:	The CVS range for Na ₂ O is 5 to 20 wt% (Table 5-3). High Na ₂ O tends to reduce glass durability and increase the likelihood of nepheline precipitation in the canister.
Li ₂ O:	The upper limit for Li ₂ O should be 3 to 4 wt% Li ₂ O to minimize possible precipitation of eucryptite (LiAlSi ₂ O ₆) from high-lithium and -alumina containing glasses. Eucryptite crystals have been known to precipitate from borosilicate glass with 16.4 to 24.6 wt% Al ₂ O ₃ and 5.9 to 6.5 wt% Li ₂ O. Na ₂ O is generally preferred for viscosity control because of Li ₂ O induced precipitation of lithium phosphates and lithium aluminum silicates (except where nepheline is more likely to precipitate than eucryptite).
CaO:	CaO may be used if it is necessary to reduce the viscosity but not the durability of glass. This situation typically occurs when the alkali components have reached their maximum acceptable limit. The CVS range for CaO is 0 to 10 wt% (Table 5-3).
MgO:	MgO should not be used as a glass or frit additive component because MgO typically increases the liquidus temperature of spinel and zirconium-containing phases, and decreases durability.
Melting Temperature:	For each composition, glass-melting temperatures should match the operating temperature of the melter (1,150 °C for the LTM).
Glass Viscosity:	The glass viscosity should be 5 Pa-s but could vary from 4 to 6 Pa-s at the operating temperature of the melter (1,150 °C).
Elec. Conductivity:	The electrical conductivity of the glass should be between 0.18 and 0.50 S/cm at 1,150 °C.
Liquidus Temperature:	As stated in the previous paragraph, the liquidus temperature limit for the LTM is 1,050 °C (100 °C less than the nominal operating temperature of the melter - 1,150 °C) to minimize spinel and zirconia precipitation in the glass and the development of sludge forming conditions in the melter.

Durability: Durability estimates are based on normalized PCT boron and sodium releases. The composition is controlled such that boron and sodium releases are limited to less than 2 g/m² over the 7-day PCT interval. This is a conservative limit designed to compensate for uncertainty in durability model prediction for glass compositions that may be outside the normal range the model. Glass compositions should not be adjusted if boron and sodium PCT releases are less than 2 g/m². In addition, glass composition should be adjusted to preclude nepheline precipitation in the canister, as determined by optical and scanning electron microscopy (SEM) and PCT boron and sodium releases from the CCC samples.

Nepheline Region: As discussed in Section 5.6, FY 1996 work indicates the nepheline region in the phase diagram is generally defined by the following limits:

1. $0.43 < \text{SiO}_2 < 0.62$
2. $\text{Al}_2\text{O}_3 < 1.67 * \text{Na}_2\text{O}$
3. $\text{Al}_2\text{O}_3 < 0.28$
4. $\text{Al}_2\text{O}_3 + 0.57 * \text{SiO}_2 > 0.43$

where Na_2O , Al_2O_3 and SiO_2 are mass fractions of each component normalized to the sum of the three components. The nepheline can be avoided by defining a minimum limit for SiO_2 such that glasses with a normalized SiO_2 content ≥ 0.62 (based on the sum of the $\text{SiO}_2 + \text{Na}_2\text{O} + \text{Al}_2\text{O}_3$) should not form nepheline in the canister.

Glass Models: The CVS first and second-order mixture models should be used to predict the viscosity, electrical conductivity and durability properties of the glass. Several models are available for liquidus temperature prediction, but these models, and especially the component coefficients in these models, need to be improved by statistical analysis of the CVS database, the DWPF (spinel) database and recent laboratory data that define the component concentration limits for spinel and zirconium containing phases in borosilicate glasses.

PNNL used these criteria to develop two separate glass compositions for each of the identified wastes, one with 25 wt% and the other with more than 25 wt% adjusted feed oxide (W'). The adjusted feed oxide (W') is defined as the weight of total non-volatile feed oxides excluding sodium and silica on an oxide basis divided by the weight of the product glass. The glass compositions and predicted properties are shown in Tables 6-2, 6-3, and 6-4 for the water and caustic washed nominal blend, water washed AZ blend and water washed 241-C-106A and 241-C-106B waste compositions, respectively.

WHC-SD-WM-TI-768
Revision 0

Table 6-2. Waste Loadings, Glass Compositions, Glass Property Predictions, and Additive Composition: Nominal Blend Waste.

Waste	NOM-1	NOM-2	NOM-3	NOMC-1	NOMC-2
Waste loading					
W'	0.35	0.25	0.30	0.30	0.25
W	0.56	0.40	0.48	0.51	0.43
Glass					
SiO ₂	0.4003	0.5011	0.4113	0.4535	0.5221
B ₂ O ₃	0.0500	0.0982	0.1093	0.0500	0.0588
Na ₂ O	0.1921	0.0913	0.1733	0.1883	0.1177
Li ₂ O	0.0100	0.0600	0.0100	0.0100	0.0521
CaO	0.0119	0.0086	0.0102	0.0119	0.0100
MgO	0.0071	0.0051	0.0060	0.0071	0.0059
Fe ₂ O ₃	0.1500	0.1076	0.1278	0.1500	0.1254
Al ₂ O ₃	0.0949	0.0681	0.0808	0.0493	0.0412
ZrO ₂	0.0216	0.0155	0.0184	0.0216	0.0181
P ₂ O ₅	0.0055	0.0039	0.0046	0.0019	0.0015
Others	0.0566	0.0406	0.0482	0.0563	0.0471
Glass Properties (Predicted)					
T _M (°C)	1,150	1,150	1,150	1,150	1,150
T _L (°C)	--	990	1,053	1,042	998
r _B (g/m ²)	0.74	0.62	1.07	1.72	1.23
r _{Na} (g/m ²)	0.37	0.33	0.47	0.77	0.81
Additives					
SiO ₂	0.7071	0.7308	0.6409	0.7447	0.7812
B ₂ O ₃	0.1127	0.1639	0.2089	0.1013	0.1019
Na ₂ O	0.1574	0.0047	0.1311	0.1336	0.0256
Li ₂ O	0.0228	0.1006	0.0191	0.0204	0.0913

Notes: Waste loadings and component concentrations are in mass fractions.

Symbols: W waste loading, W' adjusted feed oxide loading, T_M melting temperature (temperature at which glass viscosity is 5 Pa-s), T_L liquidus temperature predicted using the empirical second-order model, r_B and r_{Na} 7-day PCT normalized release of boron and sodium.

Table 6-3. Waste Loadings, Glass Compositions, Glass Property Predictions, and Additive Composition: AZ Blend Waste.

Waste	AZ-1	AZ-2	AZ-3	AZ-4	AZ-5
Waste loading					
W'	0.32	0.30	0.30	0.25	0.25
W	0.41	0.38	0.38	0.32	0.32
Glass					
SiO ₂	0.4221	0.4038	0.4296	0.5024	0.4588
B ₂ O ₃	0.0500	0.1010	0.0500	0.1049	0.0974
Na ₂ O	0.1967	0.1861	0.2114	0.0835	0.1796
Li ₂ O	0.0100	0.0100	0.0100	0.0600	0.0149
CaO	0.0033	0.0031	0.0031	0.0026	0.0026
MgO	0.0012	0.0011	0.0011	0.0009	0.0009
Fe ₂ O ₃	0.1500	0.1397	0.1397	0.1164	0.1164
Al ₂ O ₃	0.0713	0.0664	0.0664	0.0554	0.0554
ZrO ₂	0.0283	0.0264	0.0264	0.0220	0.0220
P ₂ O ₅	0.0029	0.0027	0.0027	0.0023	0.0023
Others	0.0641	0.0597	0.0597	0.0497	0.0497
Glass Properties (Predicted)					
T _M (°C)	1150	1117	1128	1150	1150
T _L (°C)	--	--	1,026	964	956
r _B (g/m ²)	1.35	2.00	2.00	0.75	2.00
r _{Na} (g/m ²)	0.51	0.65	0.71	0.36	0.79
Additives					
SiO ₂	0.6996	0.6390	0.6807	0.7262	0.6623
B ₂ O ₃	0.0833	0.1621	0.0796	0.1528	0.1419
Na ₂ O	0.2003	0.1828	0.2236	0.0331	0.1740
Li ₂ O	0.0168	0.0161	0.0160	0.0879	0.0218

Notes: Waste loadings and component concentrations are in mass fractions.

Symbols: W waste loading, W' adjusted feed oxide loading, T_M melting temperature (temperature at which glass viscosity is 5 Pa-s), T_L liquidus temperature predicted using the empirical second-order model, r_B and r_{Na} 7-day PCT normalized release of boron and sodium.

WHC-SD-WM-TI-768
Revision 0

Table 6-4. Waste Loadings, Glass Compositions, Glass Property Predictions, and Additive Composition: 241-C-106 Tank Waste.

Waste	C106A-1	C106A-2	C106A-3	C106A-4	C106B-1	C106B-2
Waste loading						
W'	0.37	0.34	0.32	0.25	0.28	0.25
W	0.85	0.80	0.75	0.58	0.45	0.41
Glass						
SiO ₂	0.3700	0.3849	0.4135	0.4806	0.3700	0.4501
B ₂ O ₃	0.0501	0.0500	0.0500	0.0831	0.1853	0.1094
Na ₂ O	0.2136	0.2209	0.1886	0.1461	0.1316	0.1311
Li ₂ O	0.0007	0.0000	0.0252	0.0402	0.0351	0.0599
CaO	0.0296	0.0278	0.0261	0.0200	0.0020	0.0018
MgO	0.0194	0.0182	0.0171	0.0132	0.0005	0.0005
Fe ₂ O ₃	0.1325	0.1248	0.1170	0.0906	0.0871	0.0781
Al ₂ O ₃	0.1375	0.1295	0.1214	0.0940	0.1604	0.1440
ZrO ₂	0.0049	0.0046	0.0044	0.0034	0.0006	0.0005
P ₂ O ₅	0.0101	0.0095	0.0089	0.0069	0.0105	0.0094
Others	0.0316	0.0298	0.0279	0.0216	0.0170	0.0152
Glass Properties (Predicted)						
T _M (°C)	1,150	1,150	1,120	1,150	1,150	1,150
T _L (°C)	--	1,011	979	927	999	877
r _B (g/m ²)	0.36	0.44	0.49	0.56	1.73	0.58
r _{Na} (g/m ²)	0.30	0.34	0.47	0.47	0.69	0.27
Additives						
SiO ₂	0.6627	0.6517	0.6995	0.7057	0.5994	0.6945
B ₂ O ₃	0.3327	0.2496	0.1997	0.1982	0.3366	0.1828
Na ₂ O	0.0000	0.0987	0.0000	0.0000	0.0000	0.0220
Li ₂ O	0.0047	0.0000	0.1008	0.0960	0.0640	0.1007

Notes: Waste loadings and component concentrations are in mass fractions.

Symbols: W waste loading, W' adjusted feed oxide loading, T_M melting temperature (temperature at which glass viscosity is 5 Pa-s), T_L liquidus temperature predicted using the second-order, r_B and r_{Na} 7-day PCT normalized release of boron and sodium.

Eleven glasses were formulated to experimentally verify the predicted properties for each glass, with two glasses representing the bounding composition limits for each waste ($W' = 25$ wt% and $W' = \text{max wt\%}$) plus one additional glass for 241-C-106A waste (Vienna et al. 1996). The properties of these glasses are listed in Tables 6-5, 6-6, and 6-7. The following sections summarize formulation and experimental results for each of the identified feed composition scenarios.

6.2.1 Glass Formulation Development for the Water Washed Nominal Blend

The water washed nominal blend is projected to have high concentrations of Al_2O_3 (16.8 wt%), Fe_2O_3 (26.6 wt%) and Na_2O (21.9 wt%), Table 6-1. Current models indicate the maximum loading of this waste should be 56 wt% in glass, based on the T_L limit. Three glasses were initially formulated to represent the likely composition range for this waste. The first glass, referred to as nom-1 with 56 wt% waste, produced a few spinel crystals at 1,150 °C and an unacceptable inventory of crystals after 24 hours at 1,050 °C (T_L limit for the LTM). The nom-2 and nom-3 glasses, with 40 and 48 wt% waste, respectively, were found to be totally free of crystals after 24 hours at 1,050 °C, satisfying the T_L constraints for this waste. These two glasses were further characterized and the results are summarized in Table 6-5.

The T_m of the nom-2 glass was measured at 1,151 °C, only 1 °C above the predicted value. Normalized boron and sodium releases from the quenched glass (0.30 and 0.24 g/m²) were found to be in reasonable agreement with the predicted values for this glass (0.62 and 0.33 g/m²). After canister centerline cooling, the nom-2 glass produced 3 vol% spinel. The normalized releases of boron and sodium from the CCC glass (0.37 and 0.47 g/m²) were only slightly higher than the baseline values for the quenched glass, indicating only a very minor decrease in durability from slow cooling. For this glass, the T_L was determined to be 1,009 °C (19 °C above the predicted value based on the second order T_L model) with spinel as the primary crystalline phase. Bright green, yellow, and light brown streaks were observed in the quenched samples of glass. These colors were caused by 10 nm spherical particles, identified by transmission electron microscopy (TEM) as silver-copper-iron alloy. These colloidal particles were sensitive to thermal conditions of the glass, disappearing from the glass after annealing at 1,050 °C and during canister centerline cooling.

The T_m of nom-3 glass was found to be 7 °C above the anticipated value of 1,150 °C. Normalized boron and sodium releases from the quenched glass (0.35 and 0.26 g/m²) were lower than predicted values for this glass (1.07 and 0.47 g/m²). This sample produced 1.8 vol% spinel and 0.2 vol% silver after CCC. Normalized releases of boron and sodium from the CCC glass (0.44 and 0.49 g/m²) were slightly higher than from the quenched sample, showing a slight decrease in durability with slow cooling. The T_L was determined to be 1,047 °C, slightly below the predicted value of 1,053 °C. Spinel was identified as the primary crystalline phase, with colored streaks similar to those found in the nom-2 glass, that disappeared after annealing for 2 hours at 1,150 °C or for 24 hours at 1,050 °C. Both glasses (nom-2 and nom-3) were deemed to have acceptable properties for the LTM, with the nom-1 glass having an unacceptably high T_L (that could lead to sludge forming conditions in

the LTM). Based on these results, the maximum waste oxide loading (W) for this composition lies between 48 and 56 wt% in glass.

Table 6-5. Waste Loading, Measured Glass Properties and Crystalline Phases in the Water Washed and Caustic Washed Nominal Blend Glasses.

Waste	Nom-2	Nom-3	Nomc-1	Nomc-2
W, wt%	40	48	51	43
W', wt%	25	30	30	25
T _m , °C	1,151	1,157	1,195	1,167
η_{1150} , Pa-s	5.0	5.4	7.6	5.7
T _L , °C	1,009	1,047	1,005	998
P	S	S	S	S
V _{C,S} , vol%	3.0	1.8	2.0	1.8
V _{C,Ag} , vol%		0.2		0.2
r _{B,Q} , g/m ²	0.3	0.35	0.5	0.41
r _{Na,Q} , g/m ²	0.24	0.26	0.64	0.42
r _{B,C} , g/m ²	0.37	0.44	0.58	0.54
r _{Na,C} , g/m ²	0.47	0.49	0.67	0.56

Notes: W waste loading, W' adjusted waste loading, T_m melting temperature, η_{1150} viscosity at 1,150 °C, T_L liquidus temperature, P primary crystalline phase, S spinel (Ni,Fe)(Fe,Cr)₂O₄, Ag silver particles detected by x-ray diffraction (XRD), V_c volume percent crystals, Q quenched glass, C canister centerline cooled glass, r_b and r_{Na} 7-day PCT normalized release of boron and sodium.

Table 6-6. Waste Loading, Measured Property Values, and Crystalline Phases in Water Washed AZ Blend Glasses.

Waste	AZ-3	AZ-5
W, wt%	38	32
W', wt%	30	25
T _m , °C	1,179	1,150
η_{1150} , Pa-s	6.8	5.1
T _L	1,048	953
P	S	S
V _{C,S} , vol%	2.0	< 1
r _{B,Q} , g/m ²	0.46	0.41
r _{Na,Q} , g/m ²	0.78	0.43
r _{B,C} , g/m ²	0.66	0.43
r _{Na,C} , g/m ²	0.71	0.54

Notes: W waste loading, W' adjusted waste loading, T_m melting temperature, η_{1150} viscosity at 1,150 °C, T_L liquidus temperature, P primary crystalline phase, S spinel (Ni,Fe)(Fe,Cr)₂O₄, Ag silver particles detected by x-ray diffraction (XRD), V_c volume percent crystals, Q quenched glass, C canister centerline cooled glass, r_b and r_{Na} 7-day PCT normalized release of boron and sodium.

Table 6-7. Waste Loading, Measured Glass Properties and Crystalline Phases in the Water Washed 241-C-106A and 241-C-106B Glasses.

Waste	C106A-2	C106A-3	C106A-4	C106B-1	C106B-2
W, wt%	80	75	58	45	41
W', wt%	34	32	25	28	25
T _m , °C	1,190	1,177	1,160	1,131	1,153
η_{1150} , Pa-s	7.5	6.5	5.6	4.3	5.1
T _L , °C	1,004	1,029	885	986	894
P	S	S	S	H, S	S
V _{C,N} , vol %	30.0	20.0	0	0	0
V _{C,S} , vol %	5.0	<1	0	<1	<2
r _{B,Q} , g/m ²	0.49	0.42	0.28	1.83	0.33
r _{Na,Q} , g/m ²	0.76	0.64	0.37	0.88	0.31
r _{B,C} , g/m ²	4.98	1.88	0.27	1.39	0.37
r _{Na,C} , g/m ²	2.61	1.16	0.46	0.77	0.50

Notes: W waste loading, W' adjusted waste loading, T_m melting temperature, η_{1150} viscosity at 1,150 °C, T_L liquidus temperature, P primary crystalline phase, S spinel (Ni,Fe)(Fe,Cr)₂O₄, Ag silver particles detected by x-ray diffraction (XRD), V_c volume percent crystals, Q quenched glass, C canister centerline cooled glass, r_b and r_{Na} 7-day PCT normalized release of boron and sodium.

6.2.2 Glass Formulation Development for the Caustic Washed Nominal Blend

The caustic washed nominal blend characteristically has less Al₂O₃ (9.6 wt%) but more Fe₂O₃ (29.2 wt%) and Na₂O (24 wt%) than the water washed blend. Based on current models, the maximum waste loading is about 53 wt% in glass, as indicated by T_L constraints. Two separate glasses were formulated from this waste, representing the bounding composition limits for this blend, Table 6-5. The first glass, referred to as nomc-1 with 51 wt% waste (W), was found to be crystal free when quenched from 1,150 °C and had a few spinel crystals and noble metal particles after 24 hours at 1,050 °C. The spinel crystals, residues from the melting reaction, dissolved after prolonged heat treatment. The nomc-2 glass with 43 wt% waste was also crystal free at 1,150 °C and after 24 hours at 1,050 °C.

The T_m of the nomc-1 glass was found to be 1,195 °C, 45 °C above the expected value based on the Arrhenius first-order viscosity model but close to the estimated value from the Fulcher first-order model (Section 5-3, Table 5-4). Normalized boron and sodium releases from quenched glass (0.5 and 0.64 g/m²) were lower than predicted (1.72 and 0.77 g/m²). Canister centerline cooled glass produced approximately 2 vol% spinel. Normalized releases of boron and sodium from CCC glass (0.58 and 0.67 g/m²) indicated a slight decrease in durability as a function of CCC cooling. The T_L of the glass was determined to be 1,005 °C (37 °C below the predicted value of 1,042 °C). Silver alloy colloidal particles precipitated from the glass after 1 hour at 1,150 °C and disappeared as in previous samples after further annealing at this temperature.

The T_m of the nomc-2 glass, at 1,167 °C, was 17 °C above the predicted value for this composition. Normalized releases of boron and sodium from quenched glass (0.41 and 0.42 g/m²) were lower than predicted (1.23 and 0.81 g/m²). Canister centerline cooled glass produced about 1.8 vol% spinel and 0.2 vol% silver, producing a slight decrease in durability (normalized boron and sodium releases of 0.54 and 0.56 g/m², respectively). The T_L was determined to be 998 °C (that matches the predicted value) with spinel as the primary phase as expected. Silver alloy streaks were observed in glasses quenched from 1,150 °C after 1 to 2 hours at this temperature. Both glasses (nomc-1 and nomc-2) were found to have acceptable properties for the LTM. From these results, it appears that the maximum oxide waste loading (W) for this composition is 51 wt% in glass.

6.2.3 Glass Formulation development for the Water Washed AZ Blend

This waste (water washed neutralized current acid waste from Tanks 241-AZ-101 and 241-AZ-102) has a considerable amount of Al₂O₃ (17.4 wt%), Fe₂O₃ (36.6 wt%) and Na₂O (19.2 wt%). The maximum waste oxide loading is projected to be 43 wt%, based on T_L limits of the glass. Five glasses (AZ-1, -2, -3, -4, and -5) were initially formulated to represent the expected composition range for this blend (Table 6-3). The first glass (AZ-1) with 41 wt% waste contained a few spinel crystals at 1,150 °C but was significantly crystallized after 24 hours at 1,050 °C. Glasses AZ-2, -3, and -5 did not show any evidence of crystallinity at 1,150 °C. The second glass (AZ-2) with 38 wt% waste was slightly crystallized (with mostly noble metals), while the third glass (AZ-3) also with 38 wt% waste and the fifth glass (AZ-5) with 32 wt% waste were crystal free after 24 hours at 1,050 °C. However, the fourth glass (AZ-4) with 32 wt% waste contained spinel at the melting temperature (1,150 °C) and also at 1,050 °C after 24 hours. Two of these glasses (AZ-3 and AZ-5) were selected for further evaluation, the results of which are summarized in Table 6-6, with symbols identical to those used in Table 6-5.

The AZ-1 and AZ-3 glasses were similar, except for slight differences in the Fe₂O₃, Al₂O₃ and Na₂O that produced crystals in the AZ-1 glass, Table 6-4. The AZ-2 glass was also similar to the AZ-3 glass, except for higher B₂O₃ and small differences in SiO₂ and Na₂O, all of which had a negative impact on the liquidus temperature of the AZ-2 glass. Glass four (AZ-4) apparently produced crystals at 1,150 °C, and again at 1,050 °C, because

of high B_2O_3 and low Na_2O content. These examples are important because they illustrate that small differences in composition can have an important effect on the crystallization behavior of the glass.

The T_m of AZ-3 glass (1,179 °C) was 51 °C above the predicted value based on the Arrhenius and Fulcher first-order viscosity models. Normalized boron release from quenched glass (0.46 g/m²) was significantly lower than the predicted value for this composition (2.0 g/m²), while the normalized sodium release from this sample (0.78 g/m²) was close to the expected value (0.71 g/m²). After CCC this glass produced 2.0 vol% spinel. Normalized boron and sodium releases from the CCC glass (0.66 and 0.71 g/m²) were similar to those from the quenched glass. The T_L was determined to be 1,048 °C (22 °C above the predicted value) with spinel the primary phase as expected. No color streaks were observed in samples of AZ-3 glass.

The T_m of AZ-5 glass was 1,155 °C, only 5 °C above the predicted value. Normalized boron and sodium releases from quenched glass were 0.41 and 0.43 g/m² and from CCC glass 0.43 and 0.54 g/m², respectively. These values were lower than the predicted release rates for this glass, 2.0 and 0.79 g/m², respectively. During canister centerline cooling, this glass formed less than 1 vol% spinel. The T_L for this composition was 953 °C (within 3 °C of the predicted value of 956 °C) with spinel again as the primary phase. Silver alloy colloidal particles formed after 1 hour in the molten glass but disappeared with further heating. Based on these data, the maximum waste loading for the water washed AZ blend (W) appears to be about 38 wt% in glass.

6.2.4 Glass Formulation Development for the Water Washed 241-C-106A Tank Waste

The Weiss analysis indicates the 241-C-106 tank waste has a considerable inventory of Al_2O_3 (16.2 wt%), Fe_2O_3 (15.6 wt%), Na_2O (25.2 wt%), and SiO_2 (31.8 wt%), Table 6-1. The maximum waste loading is projected to be 85 wt%, limited by either reduced durability (caused by nepheline precipitation during canister cooling) or by T_L (spinel crystallization in the melter). Four glasses (C106A-1,-2,-3 and-4) were initially formulated to represent the composition range for this waste. It was determined that all of these glasses were crystal free at 1,150 °C and all except C106A-1 (with 85 wt% waste loading that produced a substantial amount of spinel) were also crystal free after 24 hours at 1,050 °C. After CCC cooling, glass C106A-2 with 80 wt% waste was substantially crystallized, precipitating nepheline and spinel; glass C106A-3 with 75 wt% waste produced nepheline and less than 1 vol% spinel; while glass C106A-4 with 58 wt% waste was crystal free. Glasses C106A-2, C106A-3, and C106A-4 were further characterized and the results are summarized in Table 6-7 (with the same symbols as Table 6-5, except for H that refers to the hematite phase). Silver colloid precipitation was not observed in these glasses.

The T_m of the C106A-2 glass (1,190 °C) was 40 °C above the estimated value based on the Arrhenius first-order viscosity model but close to the value derived from the Fulcher first-order model (Table 5-4). Normalized boron and sodium releases from quenched glass

(0.49 and 0.76 g/m²) were found to be in reasonable agreement with predicted values (0.44 and 0.34 g/m²). Canister centerline cooling precipitated about 30 vol% nepheline and 5 vol% spinel and caused a substantial increase in normalized releases of boron and sodium (4.98 and 2.61 g/m²), thus signifying a significant decrease in durability (to nearly 70 percent of the allowable B limit for the DWPF EA glass). The measured T_L was 1,004 °C (slightly below the predicted value of 1,011 °C) with spinel as the primary crystalline phase.

For glass C106A-3, the T_m was 1,177 °C, 57 °C higher than predicted. Normalized releases of boron and sodium from quenched glass (0.42 and 0.64 g/m²) were close to those predicted (0.49 and 0.47 g/m²). After canister centerline cooling, 20 vol% nepheline was detected in the glass. Precipitation of nepheline caused increased normalized boron and sodium releases from the CCC glass (1.88 and 1.16 g/m²), thus indicating a significant decrease in durability (to about 26 percent of the allowable B limit for DWPF glass). The measured T_L was 1,029 °C (nearly 50 °C above the predicted value of 979 °C) with a spinel phase as the primary crystalline phase.

Glass C106A-4 had a T_m of 1,160 °C, 10 °C above the Arrhenius predicted value for this glass. Normalized boron and sodium releases for quenched glass were 0.28 and 0.37 g/m² and changed little after canister centerline cooling (0.27 and 0.46 g/m²). These values are in reasonable agreement with predicted values (0.56 and 0.47 g/m²). The T_L was determined to be 885 °C (42 °C below the predicted value) with a spinel phase as the primary crystalline phase.

Normalized boron and sodium releases from canister centerline cooled C106A-2 and C106A-3 glasses appear to be acceptable based on the WAPS 1.3 acceptance criteria defined by DWPF (two standard deviations below the mean PCT results of the EA glass, Table 5-8). Based on these results, the maximum achievable loading for 241-C-106A waste (W) appears to lie in the range from 75 to 80 wt%, but could be as low as 58 wt% if limits are imposed on the amount of nepheline in the HLW form.

6.2.5 Glass Formulation Development for the Water Washed 241-C-106B Tank Waste

The 241-C-106B (Hara) tank waste composition estimate is high in Al₂O₃ (35.6 wt%), Fe₂O₃ (19.3 wt%) and Na₂O (29.2 wt%). The maximum waste loading is estimated to be 45 wt%, limited by reduced durability (caused by nepheline precipitation during canister cooling). Two glasses, labeled C106B-1 and C106B-2, were formulated to represent the bounding conditions for this waste. Both glasses were observed to be crystal free after 24 hours at 1,050 °C. Glass C106B-1 with 45 wt% waste produced less than 2 vol% spinel after canister cooling, while glass C106B-2 with 41 wt% waste contained less than 1 vol% spinel after CCC cooling. These glasses were further characterized and the results are summarized in Table 6-7.

The T_m of C106B-1 glass was measured to be 1,131 °C, 19 °C below the predicted value. Normalized releases of boron and sodium from quenched glass were 1.83 and 0.88 g/m², that are agreeably close to the predicted values (1.73 and 0.69 g/m²). After canister centerline cooling, this glass was found to have less than 1 vol% spinel, that did not effect normalized releases of boron and sodium (1.39 and 0.77 g/m²). The T_L was determined to be 986 °C (only 13 °C below the predicted value of 999 °C) with spinel and hematite as the main phases. No evidence of silver colloid was found in the quenched glass after 1 hour at 1,150 °C, but significant streaking, typical for silver colloid, occurred after 2 hours at the same temperature. Similar streaking has also been observed in the DWPF radioactive glasses due to copper precipitation (where copper is added as a catalyst in the DWPF salt processing cell).

The T_m of C106B-2 glass was 1,153 °C, only 3 °C above the initial projection. Normalized boron and sodium releases from quenched glass (0.33 and 0.31 g/m²) and for CCC glass (0.37 and 0.5 g/m²) were reasonably close to the predicted values (0.58 and 0.27 g/m²). The measured T_L was 894 °C (slightly above the predicted value of 877 °C) with a spinel phase as the primary phase in the molten glass. The maximum waste loading (W) for this composition appears to be about 45 wt% in glass based on durability properties of the quenched sample.

6.3 GLASS VOLUME PREDICTIONS

With better spinel and nepheline models, the glass formulation models can now be used to test the bounding glass composition limits for each of the Phase I cases. Examples will be presented of the glass formulation methodology used to evaluate each of the Phase I glasses. The main purpose of this exercise is to estimate bounding conditions for each waste and the effects of retrieval sequence, blending strategy or pretreatment process on the number of canisters produced. To evaluate sensitivities, composition boundaries were evaluated by increasing the allowable liquidus temperature or by assuming that part of the Al_2O_3 and Na_2O could be removed (by caustic washing) and the glass reformulated to the maximum WOL. The impact of underestimating or overestimating the concentration of other components in the waste was also evaluated (Fe_2O_3 , Cr_2O_3 , MnO_2 , and Al_2O_3). The models that will be used include the Fulcher first-order viscosity model (Table 5-4), spinel second-order liquidus temperature model (Table 5-8), second-order B and Na PCT release models (Table 5-9) and the nepheline model (Section 5.6, Figure 5-5).

6.3.1 Bounding Composition Limits for the (Water Washed) Nominal Blend Glass

The water washed nominal blend was previously defined in Table 6-1, and the glasses developed from this waste in Tables 6-2 and 6-5. Based on the glass formulation studies to date, the optimum formulation appears to be the nom-3 glass described in Table 6-5 (with the maximum waste oxide loading (WOL) of 48 wt% in glass). The glass formulation models were used to estimate the properties of the baseline (nom-3) glass, and changes induced by:

(1) raising the WOL and allowable liquidus temperature, (2) gradually reducing the amount of Al_2O_3 and Na_2O , and (3) increasing or decreasing the amount of Fe_2O_3 , Cr_2O_3 and MnO in this waste. All of the glasses were formulated to meet a target viscosity range of 4 to 6 Pa-s in the melter (1,150 °C) and acceptable durability. The liquidus temperature of the base case (nom-3) glass is 1,047 °C, only 4 °C below the predicted value derived from the spinel model. The results are summarized in Table 6-8, with the following symbols: WOL waste oxide loading, Mass of glass (in MT), Cans showing the number of standard DWPF canisters of glass, T_L liquidus temperature of spinel, r_B and r_{Na} 7-day PCT normalized release of boron and sodium.

Table 6-8 shows the property changes that occur as the WOL and liquidus temperature limit are raised and Al_2O_3 and Na_2O concentrations are reduced in the baseline (nom-3) glass. Based on these conditions, it appears that 75 canisters of glass could be saved if the liquidus temperature limit is raised to 1,150 °C (an impractical upper limit), or 53 canisters saved if the T_L limit is raised to 1,100 °C. This may be an attractive option if a new spinel model (with ± 25 °C uncertainty) is used as a basis for reducing the liquidus temperature margin for glass formulation or if methods are developed to remedy spinel phase accumulation problems in the melter. If T_L is raised to 1,100 °C, the repository disposal fee for this waste could be reduced \$18.9 million, based on \$356,000 per canister (DOE 1995).

Table 6-8. Development of Bounding Composition Limits for the
(Water Washed) Nominal Blend Glass.

Step	WOL, wt%	Mass, MT	Cans ^c	Fe_2O_3 , wt%	Al_2O_3 , wt%	T_L , °C	r_B , g/m ²	r_{Na} , g/m ²
Nom-3 Glass	48.49	817.4	495	12.41	9.16	1,051 ^a	0.78	0.32
Incre.WOL & T_L	54.34	729.3	442	13.91	10.27	1,101 ^a	0.52	0.17
Incre.WOL & T_L	57.21	692.8	420	14.64	10.81	1,150 ^a	0.37	0.11
Red.Al 25 %	47.28	798.6	484	12.70	7.03	1,015	1.35	0.80
Red.Na	47.88	788.6	478	12.86	7.12	1,034	1.07	0.63
Red.Al 35 %	47.38	781.1	473	12.99	6.23	1,019	1.39	0.92
Red.Al 50 %	46.62	769.9	467	13.18	4.86	997	2.15 ^b	1.68
Red.Na	47.86	749.9	455	13.53	4.99	1,040	1.33	1.03
Red.Al 60 %	47.33	742.4	450	13.66	4.04	1,025	1.84	1.55
Red.Al 75 %	46.52	731.2	443	13.87	2.56	1,003	3.15 ^b	2.94
Red.Na & Li	47.73	712.7	432	14.32	2.63	1,040	1.98 ^b	1.05

^aLimited by liquidus temperature ($T_L < 1,050$ °C).

^bLimited by durability ($r_B < 2.0$ g/m²).

^cDefense Waste Processing Facility canisters (1,650 kg of glass per canister).

The glass formulation models also show that another 63 canisters of glass could be saved if the Al_2O_3 concentration is reduced by 75 percent in the waste. In this case, the minimum Al_2O_3 concentration is controlled by the durability and liquidus temperature properties of the glass, and not by nepheline precipitation in the canister (normalized $\text{SiO}_2 > 0.7118$). Since B and Na PCT release rates (r_B and r_{Na}) increase as Al_2O_3 is reduced, Na_2O and Li_2O also need to be reduced to meet the durability constraints for this glass (release rates of less than 2.0 g/m^2). The optimum glass, with 2.63 wt% Al_2O_3 , appears to have acceptable viscosity, liquidus temperature and durability properties (viscosity 5.3 Pa-s , T_L $1,040^\circ\text{C}$ and $r_B < 2.0 \text{ g/m}^2$). If 63 canisters of glass could be saved, this would translate to a repository cost savings of \$22.4 million, based on \$356,000 per canister (DOE 1995). Over this Al_2O_3 range, the WOL changes very little (only 1.5 percent) as the quantity of glass is reduced by 15 percent.

The base case (nom-3) glass contains 12.41 wt% Fe_2O_3 , 0.18 wt% Cr_2O_3 , 0.4 wt% MnO and 9.16 wt% Al_2O_3 . The liquidus temperature is projected to increase (or decrease) about 29°C for a 0.1 wt% change in Cr_2O_3 , 6°C for MnO, 2.2°C for Fe_2O_3 and 1.6°C for Al_2O_3 . Cr_2O_3 obviously has a dominate effect on liquidus temperature, while Fe_2O_3 and Al_2O_3 have roughly equivalent effects on T_L .

6.3.2 Bounding Composition Limits for the (Caustic Washed) Nominal Blend Glass

The caustic washed nominal blend is also described in Table 6-1, while the simulant glasses made from this waste are described in Tables 6-2 and 6-5. The optimum formulation appears to be the nomc-1 glass, limited by the few spinel crystals observed in this glass after 24 hours at $1,050^\circ\text{C}$. The current set of glass formulation models was used to estimate the properties of this glass, as well as changes induced by: (1) raising the WOL and liquidus temperature, (2) reducing the amount of Al_2O_3 and Na_2O and (3) increasing or decreasing the amount of Fe_2O_3 , Cr_2O_3 and MnO in this waste. All of the glasses were formulated to a meet a target viscosity range of 5 to 7 Pa-s (except the baseline nomc-1 glass which experimentally had a viscosity of 7.6 Pa-s at $1,150^\circ\text{C}$). The liquidus temperature of the nomc-1 glass is $1,005^\circ\text{C}$, about 41°C below the predicted value for this glass. The results of this modelling effort are summarized in Table 6-9, using the same nomenclature as Table 6-5.

Table 6-9. Development of Bounding Composition Limits for the (Caustic Washed) Nominal Blend Glass.

Step	WOL, wt%	Mass, MT	Cans ^c	Fe ₂ O ₃ , wt%	Al ₂ O ₃ , wt%	T _L , °C	r _B , g/m ²	r _{Na} , g/m ²
Nomc-1 Glass	51.94	688.2	417	14.74	5.66	1,046	1.03	0.74
Incre. WOL & T _L	56.44	633.3	384	16.02	6.15	1,100 ^a	0.99	0.66
Incre. WOL & T _L	59.54	600.3	364	16.90	6.49	1,150 ^a	0.91	0.59
Red. Al 25%	51.25	678.4	411	14.95	4.31	1,025	1.59	1.31
Red. Al 50%	50.54	668.7	406	15.17	2.91	1,004	2.59 ^b	2.37
Red. Na Add Li	51.27	659.2	400	15.39	2.95	1,029	2.0 ^b	1.83
Red. Al 75%	50.54	649.5	394	15.62	1.50	1,008	3.46 ^b	3.4 ^b
Add Al Red. Na Add Li	50.13	654.8	397	15.49	2.10	1,051 ^a	1.99 ^b	1.90

^aLimited by liquidus temperature ($T_L < 1,050$ °C).

^bLimited by durability ($r_B < 2.0$ g/m²).

^cDefense Waste Processing Facility canisters (1,650 kg of glass per canister).

Table 6-9 illustrates the changes that are likely to occur as the WOL and liquidus temperature are raised and Al₂O₃ and Na₂O concentrations are reduced in the nomc-1 glass. From these results, it appears that 33 canisters of glass could be saved by reformulating the nomc-1 glass to a predicted liquidus temperature limit of 1,100 °C. By reformulating the glass, approximately \$11.8 million could be saved in repository disposal costs compared to the base case nomc-1 glass. Also, 53 canisters of glass could be saved if the liquidus temperature limit (predicted T_L) is raised to 1,150 °C. Because the measured T_L is 41 °C less than the predicted T_L, a higher T_L limit might be acceptable if a better liquidus temperature model is developed.

The glass models also show that another 17 to 20 canisters could be saved by removing 50 percent of the Al₂O₃ from this waste (76 percent of the total amount of Al₂O₃ since caustic washed waste only contains 52 percent of the original amount of Al₂O₃ in the nominal blend feed). As Al₂O₃ is reduced, Na₂O and Li₂O also have to be reduced to control the durability properties of the glass. Glass formulation boundaries can be further tested by removing 75 percent of the Al₂O₃ from this waste, together with a sufficient amount of Na₂O and with a small increase in Li₂O to meet the acceptable viscosity, liquidus temperature and durability limits for this glass (viscosity 5.65 Pa-s, T_L 1,150 °C and r_B < 2.0 g/m²). This step, however, saves only three additional canisters of glass. The viscosity, liquidus temperature and durability properties of the glass are the bounding constraints, rather than the nepheline forming properties of these glasses (normalized SiO₂ concentration > 0.7325).

The nomc-1 glass contains 14.74 wt% Fe_2O_3 , 0.2 wt% Cr_2O_3 , 0.48 wt% MnO and 5.66 wt% Al_2O_3 . According to the spinel model, the liquidus temperature should increase (or decrease) about 28 °C for a 0.1 wt% change in Cr_2O_3 , 3 to 4 °C for MnO , 2 °C for Fe_2O_3 and 1.5 °C for Al_2O_3 . The liquidus temperature response is obviously dominated by Cr_2O_3 , with Fe_2O_3 and Al_2O_3 also having a significant effect.

6.3.3 Bounding Composition Limits for the AZ Blend Glass

The water washed AZ blend (blend from Tanks 241-AZ-101 and 241-AZ-102) was previously described in Table 6-1, and AZ blend glasses in Tables 6-4 and 6-6. From the glass formulation studies to date, the optimum formulation appears to be the AZ-3 glass described in Table 6-6 (with a maximum WOL of 38 wt%). The glass formulation models were used to estimate the properties of this glass, and changes induced by: (1) raising the WOL and liquidus temperature, (2) gradually reducing the amount of Al_2O_3 and Na_2O , and (3) increasing or decreasing the amount of Fe_2O_3 , Cr_2O_3 , and MnO in this waste. All of the glasses were formulated to meet a target viscosity range of 5 to 7 Pa-s in the melter (1,150 °C). The liquidus temperature of the baseline glass is 1,048 °C, 40 °C above the predicted value from the spinel model. The results are summarized in Table 6-10, using the same symbols as Table 6-5.

Table 6-10. Development of Bounding Composition Limits for the AZ Blend Glass.

Step	WOL, wt%	Mass, MT	Cans ^c	Fe ₂ O ₃ , wt%	Al ₂ O ₃ , wt%	T _L , °C	r _B , g/m ²	r _{Na} , g/m ²
AZ-3 Glass	38.55	732.05	453	13.55	6.45	1,008 ^a	1.49	0.96
Incre. WOL & T _L	43.41	650.0	396	15.26	7.26	1,101 ^a	1.46	0.84
Incre. WOL & T _L	44.79	630.1	382	15.75	7.49	1,150 ^a	0.78	0.43
Red. Al 25%	37.54	720.3	437	13.77	4.92	980	2.40 ^b	1.86
Red. Na	38.12	709.37	430	13.99	4.99	1,005 ^a	1.77	1.36
Red. Al 50%	37.07	697.47	422	14.22	3.38	977	3.05 ^b	2.71 ^b
Red. Na	37.73	685.3	415	14.22	3.38	1,008 ^a	2.16 ^b	1.90
Red. Al 40%, Red. Na	37.88	695.2	421	14.27	4.07	1,005 ^a	1.99	1.67

^aLimited by liquidus temperature (T_L predicted + 40 °C < 1,050 °C).

^bLimited by durability (r_B < 2.0 g/m²).

^cDefense Waste Processing Facility canisters (1,650 kg of glass per canister).

Table 6-10 shows the property changes that occur as the WOL and liquidus temperature are raised and Al₂O₃ and Na₂O concentrations are reduced in the AZ-3 glass. It appears that 61 canisters of glass could be saved if the liquidus temperature limit is raised to 1,150 °C and 47 canisters if the T_L limit is raised to 1,100 °C (conditions which are unacceptable because the measured T_L is 40 °C above the predicted value). This example illustrates the importance of having a reliable liquidus temperature model for prediction, but also shows the possible advantages to be gained by raising the T_L limit and reducing the liquidus temperature margin for glass formulation.

The glass models also show that only about 13 canisters of glass could be saved by removing 25 percent of the Al₂O₃ from the waste. As Al₂O₃ is reduced, the Na₂O concentration also must be reduced to control the durability characteristics of the glass. The boundaries can be further tested by removing 50 percent of the Al₂O₃ in the waste. This new glass is predicted to have unacceptable durability properties (r_B > 2.0 g/m²). An attempt was made to correct this condition by reducing the Na₂O concentration of the glass. The results show that the maximum T_L limit is reached before the durability release rates, as measured by r_B, can be brought into alignment with the target value (r_B < 2.0 g/m²). Furthermore, the final viscosity was found to be too high (7.15 Pa-s versus the targeted range of 5 to 7 Pa-s). If Na₂O or Li₂O were reduced to correct the durability problem, the glass viscosity would increase to a totally unacceptable value. The other choice would be to increase the amount of SiO₂ or Al₂O₃ in the glass, but these measures would also increase both the viscosity of the glass and the total number of canisters produced. The optimum

glass appears to one in which 40 percent of the Al_2O_3 has been removed from the waste and the Na_2O concentration has been reduced to a level where the viscosity and durability properties of the glass are marginally acceptable (viscosity of 6.97 Pa-s and $t_B < 2.0 \text{ g/m}^2$). From this analysis, approximately 22 canisters of glass could be saved by reformulating the AZ-3 glass to the edge of the viscosity and durability envelopes. This result also shows that caustic washing is only marginally effective and in all probability should not be considered for the Tanks 241-AZ-101 and 241-AZ-102 wastes if the current waste composition estimates are accurate.

The AZ-3 glass contains 13.55 wt% Fe_2O_3 , 0.17 wt% Cr_2O_3 , 0.37 wt% MnO , and 6.45 wt% Al_2O_3 . Based on the present spinel model, the liquidus temperature will increase (or decrease) about 29 °C for a 0.1 wt% increase (or decrease) in Cr_2O_3 , 4.8 °C for MnO , 2.2 °C for Fe_2O_3 , and 1.7 °C for Al_2O_3 . The liquidus temperature is most strongly affected by Cr_2O_3 , with somewhat smaller effects attributable to MnO , Fe_2O_3 , and Al_2O_3 .

6.3.4 Bounding Composition Limits for the 241-C-106A Glass

The water washed 241-C-106A waste composition is presented in Table 6-1, and potential glasses made from this waste in Tables 6-3 and 6-7. The 241-C-106A composition was derived from the Weiss analysis of the 1986 core sample, and as such is probably the most representative composition profile for this tank. Recently, a number of grab samples were taken from six different elevations in the tank (35 to 76 cm. below the top surface of the sludge). The ratio of Al to Fe in these samples generally matches the expected profile for 241-C-106A waste (while the Si concentrations were found to be only 38 percent of the expected 241-C-106A value). Because glass properties are generally controlled by the amount of Al and Fe in the waste, the 241-C-106A waste profile and glasses derived from this profile are most likely to be representative of actual Phase I glass.

The C106A-4 glass (58 wt% WOL) was chosen for initial glass formulation development because this glass did not produce nepheline in the canister centerline cooled sample. From experimental glass studies, it appears the maximum achievable waste loading could be as high as 75 to 80 wt% if 20 to 30 vol% nepheline is acceptable at the canister centerline (in which case C106A-2 and C106A-3 glasses would represent the bounding conditions for this waste). The glass formulation models were used to estimate the properties of the C106A-4 glass and changes caused by: (1) raising the WOL to the nepheline and liquidus temperature limits, (2) reducing the amount of Al_2O_3 and Na_2O , and (3) increasing or decreasing the amount of Fe_2O_3 , Cr_2O_3 , and MnO in this waste. All of the glasses were formulated to meet a target viscosity range of 4 to 6.5 Pa-s in the melter (1,150 °C). The liquidus temperature of the C106A-4 glass is 885 °C, about 38 °C below the predicted value for this glass (923 °C). The results are summarized in Table 6-11, using the same nomenclature as Table 6-5, except for the defined nepheline region (which indicates whether the glass resides in the defined nepheline region as determined by the normalized SiO_2 concentration).

Table 6-11. Development of Bounding Composition Limits for the 241-C-106A Glass.

Step	WOL, wt%	Mass, MT	Cans ^c	Fe ₂ O ₃ , wt%	Al ₂ O ₃ , wt%	T _L , °C	Neph. region	r _B , g/m ²	r _{Na} , g/m ²
C106A-4 Glass	57.22	848.6	514	9.35	9.71	923	No	0.52	0.19
Incr.WOL	68.01	713.9	432	11.12	11.54	1,031	Yes ^a	0.26	0.06
Decr.WOL	67.54	718.9	436	11.04	11.46	1,028	No ^a	0.26	0.06
Red. Al 25 %	56.45	823.6	499	9.64	7.50	933	No	0.65	0.36
Increa.WOL	70.63	658.1	399	12.06	9.39	1,036	No ^a	0.40	0.16
Red. Al 50 %	55.82	796.0	482	9.97	5.17	936	No	1.07	0.80
Incre. WOL	75.7	587.0	356	13.52	7.01	1,050	No ^a	0.64	0.38
Red. Al 75 %	55.59	762.2	462	10.41	2.70	957	No	1.93	1.79
Inc. WOL	65.27	649.2	393	12.33	3.17	986	No	1.63	1.46
Inc. WOL	81.22 ^b	521.4	316	15.22	3.95	1,008	No	1.51	1.28

^aLimited by the defined nepheline range as determined by normalized SiO₂.

^bLimited by low viscosity (3.6 Pa-S).

^cDefense Waste Processing Facility canisters (1,650 kg of glass per canister).

Table 6-11 highlights the property changes that occur as the WOL and liquidus temperature are raised and Al₂O₃ and Na₂O concentrations are reduced in the C106A-4 glass. From these results, the optimum baseline formulation produces 436 canisters of glass based on the nepheline limit (as determined by normalized SiO₂ < 0.62) (67.54 percent WOL). If the durability properties of the glass are the limiting constraint (and the volume of nepheline of secondary importance), the optimum composition would be the C106A-2 glass which produces 67 fewer canisters than the base case (369 canisters compared to 436 canisters of glass).

The glass models also show that 37 canisters of glass could be saved by removing 25 percent of the Al₂O₃ from the waste and raising the WOL to the edge of the nepheline forming envelope (70.63 percent WOL). Similarly, 80 canisters could be saved if 50 percent of the Al₂O₃ inventory is removed and the WOL again raised to the edge of the nepheline forming region (75.7 percent WOL). The final step is to assume that 75 percent of the Al₂O₃ could be removed from the waste and the WOL increased from 55.59 to 81.22 percent without exceeding the nepheline, durability or liquidus temperature limits for the glass (note that Fe₂O₃ is slightly below the CVS defined limit of 15.5 wt%). Under these conditions, 120 canisters of glass could be saved (in comparison to the base case formulation at 67.54 percent WOL). When the WOL is increased to 88.51 percent, the limiting conditions appear to be high Fe₂O (16.58 wt%) and low B₂O₃ (3 wt%, compared to the 5 wt%

minimum in the WASRD glossary definition for borosilicate glass). Thus, by reformulating the glass to the edge of the nepheline envelope and pushing the limits of the caustic washing process, it appears that 120 total canisters could be saved, reducing repository costs for this waste by \$42.7 million, based on \$356,000 per canister. All of these savings are in reference to the base case C106A-4 glass (67.54 percent WOL). The number of canisters saved varies almost linearly with the amount of Al_2O_3 removed from the waste.

The C106A-4 glass contains 9.35 wt% Fe_2O_3 , 0.21 wt% Cr_2O_3 , 0.43 wt% MnO, and 9.71 wt% Al_2O_3 . Based on the present spinel model, the liquidus temperature will increase (or decrease) about 30 °C for a 0.1 wt% increase (or decrease) in Cr_2O_3 , 5.2 °C for MnO, 1.6 to 1.9 °C for Fe_2O_3 and 0.9 °C for Al_2O_3 . The liquidus temperature is most strongly affected by Cr_2O_3 , and proportionately less so by MnO, Fe_2O_3 , and Al_2O_3 .

6.3.5 Bounding Composition Limits for the 241-C-106B Glass.

The water washed 241-C-106B composition is described in Table 6-1, and 241-C-106B glasses in Tables 6-3 and 6-7. The 241-C-106B composition was estimated from an archive sample of the 1986 core from Tank 241-C-106. Because of mixing problems during sample preparation, the 241-C-106B composition probably represents the upper bounding Al to Fe ratio in the waste. The optimum glass formulation appears to be the C106B-1 glass (45 wt% WOL). However, this glass also resides in the nepheline forming region of the $\text{Na}_2\text{O}-\text{Al}_2\text{O}_3-\text{SiO}_2$ phase diagram. While the C106B-1 glass was found to be nepheline free after canister centerline cooling, nepheline might precipitate in this glass under more favorable conditions (slower cooling).

The glass formulation models were used to estimate the properties of the C106B-1 glass and changes induced by: (1) reducing the WOL to meet the nepheline composition limit (normalized $\text{SiO}_2 > 0.62$), (2) reducing the amount of Al_2O_3 and Na_2O , and (3) increasing or decreasing the amount of Fe_2O_3 , Cr_2O_3 , and MnO in this waste. All glasses were formulated to meet a target viscosity range of 4 to 6.5 Pa-s in the melter (1,150 °C). The liquidus temperature of the C106B-1 glass is 986 °C, about 22 °C below the predicted value for this glass (1,008 °C). The results are summarized in Table 6-12, using the same nomenclature as Table 6-5, except for the defined nepheline region (which indicates whether the glass resides in the defined nepheline region as determined by the normalized SiO_2 concentration).

Table 6-12. Development of Bounding Composition Limits for the 241-C-106B Glass.

Step	WOL, wt%	Mass, MT	Cans ^b	Fe ₂ O ₃ , wt%	Al ₂ O ₃ , wt%	T _L , °C	Neph. Region	r _B , g/m ²	r _{Na} , g/m ²
C106B-1 Glass	44.43	698.1	414	8.55	15.75	1,008	Yes	1.38	0.10
Red. WOL	40.21	771.2	467	7.74	14.26	968	No ^a	0.55	0.07
Red. Al 25 %	41.28	684.7	415	8.72	12.05	957	No ^a	0.91	0.20
Red. Al 50 %	43.08	592.2	359	10.08	9.28	1,021	No ^a	2.0	0.81
Red. Al 75 %	51.19	444.7	270	13.42	6.18	1,033	No ^a	1.78	1.19

^aLimited by the defined nepheline composition range (normalized SiO₂ < 0.62).

^bDefense Waste Processing Facility canisters (1,650 kg of glass per canister).

Table 6-12 summarizes the property changes that occur as the WOL is reduced to meet the nepheline composition limit and Al₂O₃, SiO₂ and Na₂O concentrations are reduced in the C106B-1 glass. From these results, it appears that 53 more canisters would be produced if the C106B-1 glass is reformulated to avoid the nepheline forming region in the Na₂O-Al₂O₃-SiO₂ phase diagram.

The glass models also show that this effect could be reversed if 25 percent of the Al₂O₃ is removed from the 241-C-106B waste (reducing the estimated number of canisters from 467 to 414). In this case, the final glass composition would be at the edge, but outside of the nepheline forming region. Similarly, 108 canisters of glass could be saved by reducing the amount of Al₂O₃ by 50 percent (and the number of canisters from 467 to 359). The final step is to assume that 75 percent of the Al₂O₃ could be removed, in which case 270 canisters of glass would be produced from this waste. This would save altogether 197 canisters of glass (reducing the number of such canisters from 467 to 270) and \$ 70.1 million in repository storage costs.

The C106B-1 glass contains 8.55 wt% Fe₂O₃, 0.17 wt% Cr₂O₃, 0.47 wt% MnO, and 15.75 wt% Al₂O₃. The liquidus temperature, base on current models, is projected to increase (or decrease) about 29 °C for each increase (or decrease) of 0.1 wt% in Cr₂O₃, 5 °C for MnO, 1.4 °C for Fe₂O₃, and about 1 °C for Al₂O₃.

6.3.6 Summary of Results for Phase I Experimental Glasses

Table 6-13 provides a summary of the Phase I experimental glass properties. while Table 6-14 summarizes the glass volume estimates for these glasses.

Table 6-13. Properties of the Phase I Experimental Glasses.

Waste	Base glass	Base glass WOL wt%	Base glass AFO, wt%	Base glass, canisters	T _L measured	T _L predicted	Glass formulation limit
Water Washed Nominal Blend	Nom-3	48.49	30.85	495	1,047	1,051	T _L
Caustic Washed Nominal Blend	Nomc-1	51.94	31.06	417	1,005	1,046	None @ baseline composition
Water Washed AZ Blend	AZ-3	38.55	29.57	453	1,048	1,008	Nepheline limit
Water Washed 241-C-106A	Reformulated C106A-4	67.54	30.65	436	--	1,028	Nepheline limit
Water Washed 241-C-106B	Reformulated C106B-1	40.21	24.86	467	--	968	Nepheline limit

AFO = Adjusted feed oxide (weight of total non-volatile oxides excluding sodium and silica on one oxide basis divided by weight of the product glass.

CW = Caustic Washed.

WW = Water Washed.

WOL = Waste oxide loading.

Nom. Blend = Nominal Blend (80 percent Tank 241-01, 80 percent Tank 241-AZ-102, 37 percent Tank 241-C-106, 15 percent Tank 241-AY-102).

AZ Blend = Tanks 241-AZ-101/241-AZ-102 Blend.

241-C-106A = Based on Weiss core sample analysis of Tank 241-C-106 waste.

241-C-106B = Based on Hara/McCown core sample analysis of Tank 241-C-106 waste.

Canisters = Number of DWPF canisters (1,650 kg of glass per canister).

Table 6-14. Glass Volume Estimates for Phase I Experimental Glasses.

Waste	Base glass	Base glass, cans.	T _L limit @ 1,100 °C cans.	Cans w/25% Al ₂ O ₃ removed	Cans w/40% Al ₂ O ₃ removed	Cans w/50% Al ₂ O ₃ removed	Cans w/75% Al ₂ O ₃ removed
WW Nom. Blend	Nom-3	495	442	478	-	455	432 ^a
CW Nom. Blend	Nomc-1	417	384	411	-	400	397 ^{a,b}
WW AZ Blend	AZ-3	453	394	430	421 ^a	-	-
WW C106A	Reform. C106A-4	436	N/A	399	-	356	316
WW C106B	Reform. C106B-1	467	N/A	415	-	359	270 ^c

WW = Water Washed.

CW = Caustic Washed.

Nom. Blend = Nominal Blend (80 percent Tank 241-AZ-101, 80 percent Tank 241-AZ-102, 37 percent Tank 241-C-106, 15 percent Tank 241-AY-102).

AZ Blend = Tank 241-AZ-101/241-AZ-102 Blend.

241-C-106A = Based on Weiss core sample analysis of Tank 241-C-106 waste.

241-C-106B = Based on Hara/McCown core sample analysis of Tank 241-C-106 waste.

Cans = Number of DWPF canisters (1,650 kg of glass per canister).

N/A = Not applicable.

^aGlass limited by PCT B release.

^bGlass limited by T_L limit (@ 1,050 °C).

^cGlass limited by nepheline region.

The glass composition boundaries are typically established by the liquidus temperature or nepheline composition limits for each glass. Experimental glasses were formulated by Pacific Northwest National Laboratory. In several instances, experimental glasses were not formulated to the edge of the liquidus temperature or nepheline composition region and so these glasses were reformulated with the glass models to optimize the potential WOL for each composition. The glass models predict that 53 fewer canisters will be produced if the liquidus temperature limit is raised from 1,050 to 1,100 °C for the water washed nominal blend. Approximately 40 to 78 canisters also could be saved if caustic washing processes are used to remove 50 percent of the Al₂O₃ from nominal blend feed (the CW nominal blend also contains about 50 percent of the of Al₂O₃ in the nominal blend feed). The caustic washing results for the nominal blend and caustic washed nominal blend are slightly different because the glass models are non-linear and the output is often affected by the initial composition of the waste (either water or caustic washed waste). For the AZ blend, the minimum glass quantity was reached after removing 40 percent of the Al₂O₃ based on the durability properties of the glass as measured by PCT boron release. By removing 40 percent of the Al₂O₃, approximately 32 canisters of glass could be saved if the AZ blend is caustic washed. The 241-C-106A family of glasses will be discussed in the next section with the individual Phase I wastes.

6.3.7 Glass Volume Estimates for Phase I Wastes

The glass models and glass property results were used to produce the following estimates for the water washed (base case) and caustic washed Phase I wastes (Tanks 241-AZ-101, 241-AZ-102 and 241-C-106 wastes and various blends of these wastes).

Table 6-15. Glass Volume Estimates for Unblended Phase I Wastes.

Phase I Tanks	Water washed waste	Waste inventory (MT) ^a	Base glass WOL, wt%	Base glass AFO, Wt%	Cans of base ^b glass (limit)	Cans w/50 % Al ₂ O ₃ removed (limit)	Cans w/75 % Al ₂ O ₃ removed (limit)
241-AZ-101	241-AZ-101	109.46	36.08	30.26	183 (T _L)	154 (T _L)	145 (T _L)
241-AZ-102	241-AZ-102	172.99	34.40	29.29	305 (T _L)	265 (T _L)	268 (T _L)
241-C-106	241-C-106A	485.52	67.45	30.65	436 (Neph.)	356 (T _L /Neph)	316 (low vis)
Total		767.97			924	775	729

^a100 percent retrieval efficiency.

^bDWPF canisters of glass limited by T_L liquidus temperature or nepheline.

Phase I Blend Compositions

Nominal Blend: $0.8 * 241\text{-AZ-101} + 0.8 * 241\text{-AZ-102} + 0.37 * 241\text{-C-106} + 0.15 * 241\text{-AY-102}$

AZ Blend: $241\text{-AZ-101} + 241\text{-AZ-102}$

AYC Blend: $241\text{-C-106} + 241\text{-AY-102}$

AYC-A Blend: $241\text{-AY-102} + 0.1625 * 241\text{-C-106}$

AYC-B Blend: $241\text{-AY-102} + 0.4875 * 241\text{-C-106}$

AZC-A Blend: $\text{AZ Blend} + 0.4875 * 241\text{-C-106}$

AZC-B Blend: $\text{AZ Blend} + 0.1625 * 241\text{-C-106}$

Table 6-16. Glass Volume Estimates for Various Phase I Blends.

Phase I blends	Water washed waste	Waste inventory (MT)	Cans ^a of base glass (limit) ^b	Cans w/50% Al ₂ O ₃ removed (limit) ^b	Cans w/75% Al ₂ O ₃ removed (limit) ^b
Nominal	Nominal	396.3	495 (T _L)	455 (T _L)	432 (T _L)
AZ Blend	AZ Blend	282.52	453 (neph.)	421 (Dur.w/40% Al ₂ O ₃ removed)	N/A
AYC	241-AY-102/ 241-C-106A	514.55	554 (neph.)	397 (neph.)	343 (T _L)
AYC-A	241-AY-102/ 241-C-106A	116.23	139 (neph.)	99 (T _L)	96 (T _L)
AYC-B	241-AY-102/ 241-C-106A	281.36	333 (neph.)	220 (neph.)	211 (durable)
AZC-A	241-AZ-101/ 241-AZ-102/ 241-C-106A	479.83	600 (neph.)	477 (T _L)	461 (T _L)
AZC-B	241-AZ-101/ 241-AZ-102/ 241-C-106A	315.14	450 (neph.)	372 (T _L)	365 (T _L)

^aDWPF canisters of glass.

^bGlass formulation limit (T_L liquidus temperature limit, neph. nepheline composition limit, durability limit).

If caustic washing processes are employed, the glass models indicate 80 canisters could be saved by removing 50 percent of the Al₂O₃ from Tank 241-C-106 waste and 120 canisters if 75 percent of the Al₂O₃ is removed from this waste. Caustic washing processes appear to be less desirable for Tanks 241-AZ-101 and 241-AZ-102 wastes, producing 29 and 40 fewer canisters, respectively. These estimates are based on the physical and chemical properties of the glass. Caustic washing studies will be necessary to determine whether it is possible to remove this much aluminum from the Phase I wastes. Also, it is unclear whether the TWRS safety basis could be modified in time to allow enough tank space for a caustic washing option for these wastes.

If Tanks 241-AZ-101/241-AZ-102 wastes are caustic washed, approximately 419 canisters could be produced from the residual waste (Table 6-15), compared to 421 canisters from the caustic washed AZ blend (Table 6-16). The total amount of glass derived from Tanks 241-AZ-101 and 241-AZ-102 waste (488 canisters base case) is slightly

more than predicted for the AZ blend (453 canisters). Again, this is an instance where the nonlinear glass models have produced slightly different results depending on the initial composition of the waste.

Glass property data indicate the maximum waste oxide loading of the nominal blend feed is 48.5 wt% total oxides or 30.9 wt% on an adjusted feed oxide (AFO) basis. The AFO basis is defined as the weight of total non-volatile feed oxides excluding sodium and silica on an oxide basis divided by weight of the product glass. If the Phase I wastes are processed separately, the average waste oxide loading is 30.1 wt% AFO. This indicates that little benefit will be gained from blending in terms of reducing the overall volume of glass. However, there may be other benefits from blending, including reducing the concentration of problem components (such as silver) and reducing the number of glasses and process conditions that need to be qualified.

Current estimates do not include any allowance for operating margin contingency (to compensate for uncertainties in the waste or glass composition analysis). If it is assumed that production glasses are formulated to 85 percent of the maximum waste loading, the average waste loading for the tank by tank and nominal blend feed are 25.6 and 26.3 percent AFO, respectively. For the tank by tank feed, this results in 957,000 kg of glass (580 canisters) for the minimum Phase I feed of 245,000 kg OFT or 1,800,000 kg of glass (1,100 canisters) for the maximum feed of 465,000 kg OFT.

Based on these models and glass property data, the maximum waste loading for the Phase I glasses varies from 29.76 to 30.65 wt% on an adjusted feed oxide basis. This indicates there is a 17 percent contingency or operating margin between the maximum theoretical waste loading and the minimum 25 percent adjusted feed oxide basis specified in the RFP. The 25 percent AFO appears to be conservative and an achievable value for Phase I.

6.3.8 Bounding Effects of Al_2O_3 and Fe_2O_3 in Glass

Sensitivity analyses were also performed to evaluate the effects of feed composition and more aggressive caustic washing processes for these wastes. The purposes of this analysis is to provide feedback to the pretreatment and waste retrieval functions, and to identify high priority issues to be addressed in any future waste characterization activities.

The most important components from a glass making perspective are Al_2O_3 and Fe_2O_3 . Because glass volume predictions are important to overall project cost, a sensitivity study was performed to assess the likely impact of varying the amount of Al_2O_3 and Fe_2O_3 in the Tanks 241-AZ-101 and 241-C-106 wastes. Two glasses were chosen for this investigation, including high waste loaded glasses developed from Tanks 241-AZ-101 and 241-C-106 wastes, which represent the bulk of the Phase I wastes. In this study, glass composition boundaries were systematically investigated by varying the amount of Al_2O_3 and Fe_2O_3 in the

waste from -75 to +100 percent of the nominal inventory and formulating the optimum glasses for each composition.

The results are summarized in Tables 6-17 and 6-18, and in Figures 6-1 through 6-8 for Tank 241-AZ-101 waste. Figure 6-1 shows the effect of varying the amount of Al_2O_3 in Tank AZ-101 based on three different glass formulation constraints. The CVS single component glass limits represent the bounding compositions used in the CVS glass formulation study (15 wt% Fe_2O_3 and 17 wt% Al_2O_3), while the liquidus temperature and nepheline limits represent the glass property limits for avoiding sludge accumulation problems in the melter and for meeting acceptable durability conditions at the canister centerline, respectively. Other properties such as viscosity and durability as measured by boron and sodium release are also important, but only rarely become the limiting constraints for glass formulation. In Figure 6-2, glass model predictions and CVS single component glass limits are presented as a function of the amount of Al_2O_3 in Tank 241-AZ-101 waste. Glass model predictions are based on the minimum volume of glass consistent with the liquidus temperature and nepheline limits.

In the region from -75 to 0 percent change in the nominal Al_2O_3 inventory, the liquidus temperature limit controls the volume of glass, while the nepheline limit controls the composition of the glass from 0 to 100 percent increase in the nominal Al_2O_3 inventory. The CVS single component glass limits tend to be overly optimistic with respect to glass model predictions for this waste. Based on the results in Table 7-4, the CVS single component limits tend to underpredict the actual amount of glass by 30 to 60 percent in the high Al_2O_3 range and 5 to 30 percent in the lower Al_2O_3 (and higher Fe_2O_3) concentration range. If CVS multi-component limits were also used ($\text{Fe}_2\text{O}_3 + \text{Al}_2\text{O}_3 + \text{ZrO}_2 + \text{others} \leq 0.24$), glass volume estimates would be closer to glass model predictions for this waste.

Table 6-17. Glass Volume as a Function of the Amount of Al_2O_3 in Tank 241-AZ-101 Waste.

Al_2O_3 , % change in mass	HLW glass model, cans	CVS glass limits, cans (limit)	Liq. temp. limit, cans	Neph. limit, cans	Al_2O_3 conc. in glass, ^a wt%	Fe_2O_3 conc. in glass, ^a wt%	Glass vol % change ^a
-75	145	139 (Fe)	145	136	2.64	14.4	-20.8
-50	154	139 (Fe)	154	150	4.97	13.54	-15.8
0	183	139 (Fe)	183	182	8.35	11.38	0
25	211	139 (Fe)	189	211	9.08	9.9	15.3
50	238	139 (Fe)	204	238	9.67	8.78	30.0
75	270	158 (Al)	219	270	9.95	7.75	47.5
100	286	180 (Al)	229	286	10.71	7.30	56.3

^aGlass composition based on the minimum volume of glass consistent with the liquidus temperature and nepheline limits (i.e., high-level waste glass model predictions).

Table 6-18. Glass Volume as a Function of the Amount of Fe_2O_3 in Tank 241-AZ-101 Waste.

Fe_2O_3 , % change in mass	HLW glass models, cans	CVS glass limits, cans	Liq. temp. limit, cans	Neph. limit, cans	Al_2O_3 conc. in glass, ^a wt%	Fe_2O_3 conc. in glass, ^a wt%	Glass vol % change ^a
-75	174	90(Al)	137	174	8.8	3.0	-4.9
-50	176	90(Al)	147	176	8.63	5.88	-3.8
-25	180	104(Fe)	161	180	8.50	8.68	-1.6
0	183	139(Fe)	183	182	8.35	11.38	0
25	202	174(Fe)	202	196	7.60	12.95	10.4
50	224	209(Fe)	224	210	6.84	13.99	23.4
75	241	244(Fe)	241	232	6.36	15.16	31.7
100	261	278(Fe)	261	250	5.86	15.98	42.7

^aGlass composition based on the minimum volume of glass consistent with the liquidus temperature and nepheline limits (i.e., high-level waste glass model predictions).

Figure 6-1. Effect of Liquidus Temperature, Nepheline, and Composition Variability Study Limits on Volume of Glass versus Al_2O_3 in Tank 241-AZ-101 Waste.

Figure 6-2. Effect of High-Level Waste Glass Model and Composition Variability Study Limits on Volume of Glass versus Al_2O_3 in Tank 241-AZ-101 Waste.

Figure 6-3. Al_2O_3 and Fe_2O_3 Concentrations in Glass versus Al_2O_3 in Tank 241-AZ-101.

Figure 6-4. Change in Volume of Glass versus Percent Change of Al_2O_3 in Tank 241-AZ-101 Waste.

Figure 6-5. Effect of Liquidus Temperature, Nepheline, and Composition Variability Study Limits on Volume of Glass versus Fe_2O_3 in Tank 241-AZ-101 Waste.

Figure 6-6. Effect of High-Level Waste Glass Model and Composition Variability Study Limits on Volume of Glass versus Fe_2O_3 in Tank 241-AZ-101 Waste.

Figure 6-7. Fe_2O_3 and Al_2O_3 Concentrations in Glass versus Fe_2O_3 in Tank 241-AZ-101 Waste.

Figure 6-8. Change in Volume of Glass versus Percent Change of Fe_2O_3 in Tank 241-AZ-101 Waste.

The Al_2O_3 and Fe_2O_3 glass compositions are shown as a function of the Tank 241-AZ-101 Al_2O_3 inventory in Figure 6-3. Tank 241-AZ-101 glass limits change from nepheline to liquidus temperature control at the base case glass composition (8.35 wt% Al_2O_3 and 11.38 wt% Fe_2O_3). This transition is important because it shows that nepheline is the limiting property above the Al_2O_3 to Fe_2O_3 ratio of 0.74 in the glass (and Tank 241-AZ-101 waste), while liquidus temperature is the limiting property below this ratio. Figure 6-4 shows the relative change in the number of canisters as a function of the amount of Al_2O_3 in Tank 241-AZ-101. The overall glass volume increases 56 percent with a 100 percent increase in the amount of Al_2O_3 in this waste, but decreases only 21 percent with a 75 percent decrease of Al_2O_3 . Under nepheline control, the glass volume increases strongly with the amount of Al_2O_3 in the waste, while the volume decreases only about one-half as much under liquidus temperature control. Therefore, caustic washing appears to be only about 50 percent effective (in terms of glass volume reduction) in the liquidus temperature (or iron solubility controlled) region of the composition envelope (below the Al_2O_3 to Fe_2O_3 ratio of 0.74 for Tank 241-AZ-101 waste).

Figures 6-5 through 6-8 show the reverse effect of varying the Fe_2O_3 content of Tank 241-AZ-101 waste. The CVS single component, liquidus temperature, and nepheline limits are shown as a function of the amount of Fe_2O_3 in the waste in Figure 6-5. Once again, the CVS single component limits are overly conservative (tend to underpredict the volume of glass) in the low Fe_2O_3 (and high Al_2O_3) range, but are in better agreement with the glass models in the high Fe_2O_3 range (150 to 200 percent of the baseline inventory). The baseline composition serves as the dividing point between the nepheline and liquidus temperature control (with the liquidus temperature control region being located to the right or in the high Fe_2O_3 glass composition region). The HLW glass model predictions and CVS single component glass limits are presented as a function of Fe_2O_3 composition in Figure 6-6. Glass predictions are based on the minimum volume of glass consistent with the nepheline and liquidus temperature limits. In this case, the CVS single component glass limits tend to underpredict the volume of glass by 30 to 60 percent in the low Fe_2O_3 region, and more closely agree with the glass models and actually overpredict the amount of glass by 6.5 percent at the extreme end of the Fe_2O_3 composition envelope (200 percent of the baseline inventory).

The predicted Fe_2O_3 and Al_2O_3 concentrations in glass are shown as a function of the Tank 241-AZ-101 Fe_2O_3 inventory in Figure 6-7. The transition between nepheline and liquidus temperature control once again occurs at the baseline glass composition (8.35 wt% Al_2O_3 and 11.38 wt% Fe_2O_3), with the nepheline region being to the left of the baseline composition and the liquidus temperature controlled region to the right. In Figure 6-8, the relative glass volume is shown as a function of Tank 241-AZ-101 Fe_2O_3 concentration. According to current models, the glass volume increases 43 percent with a 100 percent increase in the amount of Fe_2O_3 in this waste, but only decreases 4.9 percent with a 75 percent decrease of Fe_2O_3 . Under liquidus temperature control, the glass volume increases strongly with the amount of Fe_2O_3 in the waste, but decreases only slightly with a diminishing amount of Fe_2O_3 under nepheline composition control. If the mass of all components except Al_2O_3 or Fe_2O_3 is held constant, the volume of Tank 241-AZ-101 glass is

projected to increase 4.3 and 5.6 percent with a 10 percent increase in the amount of Fe_2O_3 or Al_2O_3 , respectively, and decrease 0.65 and 2.8 percent with a decrease of 10 percent in the amount of Fe_2O_3 or Al_2O_3 in this waste. These figures can be used to estimate the cost of over-estimating or under-estimating the mass of these components (based on a repository disposal fee of \$356,000 per canister). These results simulate the effect of reducing Al_2O_3 by using the caustic washing process or inadvertently adding Al_2O_3 by precipitation from the liquid phase.

The glass modelling results for Tank 241-C-106 waste are summarized in Tables 6-19 and 6-20, and in Figures 6-9 through 6-16. Figure 6-9 shows the effect of varying the amount of Al_2O_3 in Tank 241-C-106 based on the CVS single component glass, liquidus temperature and nepheline limits for HLW glass. In Figure 6-10, the HLW glass volume and CVS single component glass limits are shown as a function of the amount of Al_2O_3 in Tank 241-C-106. Glass model predictions are based on the minimum volume of glass consistent with the liquidus temperature and nepheline limits for this glass. The nepheline and liquidus temperature limits evidently converge in the range from -25 to -50 percent of the nominal Al_2O_3 inventory, while low viscosity is the limiting constraint at -75 percent Al_2O_3 . The nepheline forming characteristics of the glass control the formulation results in the range from 0 to 100 percent increase in the nominal Al_2O_3 inventory. CVS single component glass limits once again tend to underpredict the actual amount of glass by 25 to 35 percent in the range from -25 to +50 percent of the baseline Al_2O_3 inventory.

Glass Al_2O_3 and Fe_2O_3 concentrations are shown as a function of Al_2O_3 inventory in Figure 6-11. The crossover from nepheline to liquidus temperature/nepheline control occurs at -25 percent of the nominal Al_2O_3 inventory (9.39 wt% Al_2O_3 and 12.06 wt% Fe_2O_3 or Al_2O_3 to Fe_2O_3 ratio of 0.77). This transition is interesting because it occurs in the same composition region as the (nepheline to liquidus) transition in Tank 241-AZ-101 glasses. In Figure 6-12, the relative glass volume is depicted as a function of tank C-106 Al_2O_3 composition. The glass volume increases nearly 47 percent with a 100 percent increase in the amount of Al_2O_3 in the waste, and decrease 18.3 percent with a 50 percent decrease of Al_2O_3 . Under nepheline control, the volume of glass increases sharply with the amount of Al_2O_3 in the waste, and decreases at a slightly smaller rate as the amount of Al_2O_3 is reduced in the waste.

Table 6-19. Glass Volume as a Function of the Amount of Al_2O_3 in Tank C-106 Waste.

Al_2O_3 , % change in mass	HLW glass models, cans	CVS glass limits, cans (limit)	Liq. temp. limit, cans	Neph. limit, cans	Al_2O_3 conc. in glass, ^c wt. %	Fe_2O_3 conc. in glass, ^c wt. %	Glass vol% change ^c
-75	316 ^a	321 (Fe)	^b	^b	3.95	15.22	-27.5
-50	356	321 (Fe)	356	356	7.01	13.52	-18.35
-25	399	321 (Fe)	400	399	9.39	12.06	-8.49
0	436	321 (Fe)	419	436	11.45	11.03	0
25	496	367 (Al)	480	496	12.59	9.7	13.76
50	546	440 (Al)	506	546	13.73	8.82	25.23
75	590	514 (Al)	558	590	14.81	8.16	35.32
100	640	587 (Al)	621	640	15.59	7.51	46.79

^aGlass composition limited by low viscosity

^bLiquidus temperature could not be increased to target level without adding Al_2O_3 .

^cGlass composition based on the minimum volume of glass consistent with the liquidus temperature and nepheline limits (i.e., high-level waste glass model predictions).

Table 6-20. Glass Volume as a Function of the Amount of Fe_2O_3 in Tank 241-C-106 Waste.

Fe_2O_3 , % change in mass	HLW glass models, cans	CVS glass limits, cans (limit)	Liq. temp. limit, cans	Neph. limit, cans	Al_2O_3 conc. in glass, ^a wt. %	Fe_2O_3 conc. in glass, ^a wt. %	Glass vol% change ^a
-75	402	294 (Al)	390	402	12.43	2.99	-7.80
-50	409	294(Al)	395	409	12.21	5.88	-6.19
-25	418	294(Al)	404	418	11.95	8.63	-4.13
0	436	321(Fe)	419	436	11.45	11.03	0
25	451	401(Fe)	451	442	11.08	13.34	1.38
50	497	481(Fe)	497	450	10.04	14.52	13.99
75	575	561(Fe)	575	480	8.68	14.64	31.88
100	624	642(Fe)	624	480	8.0	15.42	43.10

^aGlass composition based on the minimum volume of glass consistent with the liquidus temperature and nepheline limits (i.e., high-level waste glass model predictions).

Figure 6-9. Effect of Liquidus Temperature, Nepheline, and Composition Variability Study Limits on Volume of Glass versus Al_2O_3 in Tank 241-C-106 Waste.

Figure 6-10. Effect of High-Level Waste Glass Model and Composition Variability Study Limits on Volume of Glass versus Al_2O_3 in Tank 241-C-106 Waste.

Figure 6-11. Al_2O_3 and Fe_2O_3 Concentrations in Glass versus Al_2O_3 in Tank 241-C-106 Waste.

Figure 6-12. Change in Volume of Glass versus Percent Change of Al_2O_3 in Tank 241-C-106 Waste.

Figure 6-13. Effect of Liquidus Temperature, Nepheline, and Composition Variability Study Limits on Volume of Glass versus Fe_2O_3 in Tank 241-C-106 Waste.

Figure 6-14. Effect of High-Level Waste Glass Model and Composition Variability Study Limits on Volume of Glass versus Fe_2O_3 in Tank 241-C-106 Waste.

Figure 6-15. Fe_2O_3 and Al_2O_3 Concentrations in Glass versus Fe_2O_3 in Tank 241-C-106 Waste.

Figure 6-16. Change in Volume of Glass versus Percent Change of Fe_2O_3 in Tank 241-C-106 Waste.

Figures 6-13 through 6-16 show the effect of varying the amount of Fe_2O_3 in Tank 241-C-106 waste. In Figure 6-13, the CVS single component glass, liquidus temperature and nepheline limits are presented as a function of Fe_2O_3 . Glass model predictions and CVS single component glass limits are plotted as a function of Fe_2O_3 in Figure 6-14. In this case, the CVS single component glass limits tend to underpredict the volume of glass by 35 to 40 percent in the low Fe_2O_3 region, but produce better results in the higher Fe_2O_3 region of the glass composition space. The Fe_2O_3 and Al_2O_3 glass compositions are shown in Figure 6-15, with the crossover from nepheline to liquidus temperature control occurring in the region between 100 and 125 percent of the baseline Fe_2O_3 inventory (Al_2O_3 to Fe_2O_3 ratio of 0.83 to 1.03). In the previous examples, the nepheline to liquidus transition occurred in the region defined by Al_2O_3 to Fe_2O_3 ratios of 0.74 to 0.77. The difference in this case might be related to the radical shift in first and second-order liquidus temperature model predictions over this composition range (C106A-3 and C106A-4 glasses in Table 5-9). The first and second-order models tend to underpredict the liquidus temperature (by 30 to 50 °C) in the region where Al_2O_3 and Fe_2O_3 concentrations are 12.5 to 13 wt% in glass, and overpredict liquidus temperature for the same family of glasses (by 40 to 60 °C) where the Al_2O_3 and Fe_2O_3 concentrations have been reduced to 9 to 9.5 wt%. With this level of uncertainty, additional data are needed to more accurately predict the transition from nepheline to liquidus temperature control for the Tank 241-C-106 family of Fe_2O_3 rich glasses.

In Figure 6-16, the relative glass volume is shown as a function of Tank 241-C-106 Fe_2O_3 concentration. Based on current projections, the glass volume increases 43 percent with an increase of 100 percent in the amount of Fe_2O_3 in the waste, but decreases only 8.5 percent with a 75 percent decrease in Fe_2O_3 . Again, as in the case of Tank 241-AZ-101 waste, the volume of glass increases strongly with the amount of Fe_2O_3 , but decreases only marginally under nepheline control as the amount of Fe_2O_3 is reduced in the waste. If the mass of all components except Al_2O_3 or Fe_2O_3 is constant, the volume of C-106 glass should increase 4.3 and 4.6 percent with a 10 percent increase in the amount of Fe_2O_3 or Al_2O_3 , respectively, and decrease 1.1 and 3.6 percent with a decrease of 10 percent in the amount of Fe_2O_3 or Al_2O_3 in this waste. These figures can also be used to estimate the cost of over-estimating or under-estimating the mass of these components (based on a repository disposal fee of \$356,000 per canister).

6.3.9 Glass-Limiting Components in the Waste

Tank core samples have been commonly used as a primary basis for estimating the composition of most HLW at the Hanford Site. Recently, a study was conducted to evaluate the statistical variability of samples from Tanks 241-BX-107, 241-T-107, 241-C-109, 241-C-110, and 241-C-112 (Jensen 1994). The results generally indicate that 2 to 4 cores may be required at an average cost of \$1 million per core for the analysis of most components to be within 100 percent of the mean value in each tank (95 percent confidence level). Statistical methods also have been used to assess homogeneity errors in SSTs with replicate samples (Jensen and Liebetrau 1988). Such errors appear to be caused by vertical

and horizontal inhomogeneities in the tank, which can seriously affect the accuracy of any sampling program. Tank-by-tank inventories are needed to define the retrieval sequence, downstream processing, pretreatment, and TWRS flowsheet for the processing of the HLW inventory.

Glass models can be used to estimate the most important components in the waste and the composition accuracy required for glass formulation purposes. These predictions are based on the minimum change (in waste concentration) needed to induce a 50 °C increase or decrease in the output of the second-order liquidus temperature model or a 5 percent change in the nepheline limit for Tanks 241-AZ-101 and 241-C-106 glasses. The results are summarized in Tables 6-21, 6-22, 6-23, and 6-24.

Table 6-21. Tank 241-AZ-101 Base Case Glass Composition.

Component	Frit (kg)	Waste (kg)	Tank 241-AZ-101 glass (kg)	Tank 241-AZ-101 glass (wt%)
SiO ₂	125,000	3,017	128,017	42.3
B ₂ O ₃	18,000	230	18,230	6.0
Na ₂ O*	37,000	14,605	51,605	17.0
Li ₂ O	9,000	37	9,037	3.0
CaO	0	821	821	0.3
MgO	0	244	244	0.1
Fe ₂ O ₃	0	34,444	34,444	11.4
Al ₂ O ₃	0	25,282	25,282	8.4
ZrO ₂	0	11,445	11,445	3.8
Cr ₂ O ₃	0	407	404	0.1
MnO	0	1,152	1,152	0.4
NiO	0	1,367	1,367	0.5
P ₂ O ₅	0	1,545	1,545	0.5

*The fraction of Na₂O in waste will vary with washing efficiency.

Table 6-22. Waste Concentration Change Needed to Induce a 50 °C Change in Liquidus Temperature (T_L) or 5 Percent Change in the Nepheline Limit for Tank 241-AZ-101 Glass.

Component	Source (% frit/% waste)	Component contribution to T_L , °C	Percent change needed for 50 °C increase or decrease in T_L	Percent change needed for 5% change in nepheline limit
SiO ₂	Frit 97.64%/Waste 2.36%	-244.17	867	339
B ₂ O ₃	Frit 98.74%/Waste 1.26%	115.24	3,400	
Na ₂ O	Frit 71.70%/Waste 28.30%	114.45	154	85.9
Li ₂ O	Frit 99.59%/Waste 0.41%	-23.25	5,240	
CaO	Waste 100%	10.64	470	
MgO	Waste 100%	-2.97	1,683	
Fe ₂ O ₃	Waste 100%	358.96	13.9	
Al ₂ O ₃	Waste 100%	369.93	13.5	49.6
ZrO ₂	Waste 100%	73.16	68.3	
Cr ₂ O ₃	Waste 100%	48.32	103.4	
MnO	Waste 100%	89.98	55.6	
NiO	Waste 100%	15.95	313.5	
P ₂ O ₅	Waste 100%	15.91	314.3	

Table 6-23. Tank 241-C-106 Base Case Glass Composition.

Component	Frit (kg)	Waste (kg)	241-AZ-101 glass (kg)	241-AZ-101 glass (wt%)
SiO ₂	140,000	161,960	301,960	41.9
B ₂ O ₃	64,300	50	64,350	8.9
Na ₂ O	0	102,926	102,926	14.3
Li ₂ O	18,000	0	18,000	2.5
CaO	0	17,720	17,720	2.5
MgO	0	11,596	11,596	1.6
Fe ₂ O ₃	0	79,381	79,381	11.0
Al ₂ O ₃	0	82,382	82,382	11.5
ZrO ₂	0	2,956	2,956	0.4
Cr ₂ O ₃	0	1,531	1,531	0.2
MnO	0	3,105	3,105	0.4
NiO	0	1,277	1,277	0.2
P ₂ O ₅	0	6,033	6,033	0.8

Table 6-24. Waste Concentration Change Needed to Induce a 50 °C Change in Liquidus Temperature (T_L) or 5 Percent Change in the Nepheline Limit for Tank 241-C-106 Glass.

Component	Source (% Frit/% Waste)	Component contribution to T_L , °C	Percent change reqd. for 50 °C increase or decrease in T_L	Percent change reqd. for 5% change in nepheline limit
SiO ₂	Frit 46.36%/Waste 53.64%	-251.25	37	15.0
B ₂ O ₃	Frit 99.92%/Waste 0.08%	171.07	-3,650	
Na ₂ O	Waste 100%	96.02	52	24.9
Li ₂ O	Frit 100%	-19.48	257	
CaO	Waste 100%	86.41	58	
MgO	Waste 100%	12.49	400	
Fe ₂ O ₃	Waste 100%	344.77	14.5	
Al ₂ O ₃	Waste 100%	366.50	13.6	31.1
ZrO ₂	Waste 100%	7.94	630	
Cr ₂ O ₃	Waste 100%	79.78	62.7	
MnO	Waste 100%	101.97	49.0	
NiO	Waste 100%	6.27	798	
P ₂ O ₅	Waste 100%	26.13	191.4	

These models indicate that only two components (Fe₂O₃ and Al₂O₃) need to be known with reasonable precision (to within 15 percent or their true values) to estimate the amount of glass likely to be produced from Tank 241-AZ-101 waste, and four components (SiO₂, Na₂O, Fe₂O₃ and Al₂O₃) to estimate the volume of glass from Tank 241-C-106 waste. Other components from a glass formulation perspective that need to be known (to within 100 percent of their true values) are Na₂O, MnO, and ZrO₂ for Tank 241-AZ-101 waste and Cr₂O₃, MnO, and CaO for Tank 241-C-106 waste. While it may be desirable to obtain as much information as possible, the volume and composition of most glasses can be reasonably determined from Fe₂O₃ and Al₂O₃ composition data, together with some knowledge of Cr₂O₃, MnO, noble metals (Ag₂O, PdO, Rh₂O₃, and Ru₂O₃), and the semi-soluble components in the waste (P₂O₅ and SO₄).

6.4 BALANCE OF PROCESS

Some components may have a significant effect on the operability of the vitrification facility. This section provides a discussion of feed processability issues unrelated to glass production estimates. Offgas and emissions control systems could be affected by the amount of mercury, cadmium, and other volatile components in the waste while chlorides and sulfates could contribute to the general corrosion of offgas system components. Criticality controls will undoubtedly be effected by actinides in the waste. In addition to other considerations, nitric oxide emissions from the Phase I immobilization facility could have a effect on allowable release rates from the Hanford Site. These issues will be addressed in this section of the report.

6.4.1 Mercury

Mercury was occasionally added as mercuric nitrate to uranyl nitrate feedstock in the REDOX and PUREX processes to suppress the volatility of ^{131}I when relatively short-cooled fuels were dissolved. Records indicate that mercury was used from 1952 through 1967, predominately in the PUREX process. During this period, mercury was added to the dissolvers about 25 percent of the time at a rate of approximately 0.77 kg of mercuric nitrate per 6 tons of uranium. Altogether, the total mercury inventory at the Hanford Site is estimated to be in the range of 2,000 kg, with 88 percent in the high-level PUREX waste and the balance in the high-level REDOX waste (Wagner 1992, Allen 1976). The mercury concentration varies from 830 to 1,400 p/M in the high-level PUREX sludge. An analysis of sludge samples from Tank 241-AY-102 shows 40 p/M of mercury (Voogd 1988). Samples from Tank 241-C-106 suggest the mercury concentration is between 42 to 327 p/M (McCown 1988). On a dried solids basis, as much as 0.017 wt% mercury (173 p/M) could be present in the combined Phase I feed from Tanks 241-C-106 and 241-AY-102.

Traceable amounts of copper, lead, and mercury have been found in many of the SSTs, principally those with high-level B Plant and evaporator bottoms wastes. Nine of 12 SST farms were included in the sample population (all except Tank Farms AX, T, and TY). Mercury concentrations varied from 5 to 800 p/M, with most samples being in the range of 30 to 350 p/M. If the sludge-washing calculations for Tank 241-AY-102 are considered typical, melter feeds from the SSTs could contain as much as 75 to 875 p/M mercury.

A small amount of mercuric nitrate also was used in the plutonium reclamation facility to catalyze the dissolution of plutonium-aluminum alloy scrap. This usage was small between 1973 and 1976 when the waste was sent to the 200 West Area Tank Farms.

The solubility product of mercuric hydroxide is approximately 10^{-22} , which means that at a pH of 10, the mercuric ion concentration in solution is only 10^{-14} molar. The sludge-washing partitioning factor for mercuric iodide is assumed to be zero (Higley 1988). Given the extremely low solubility of mercuric oxides under alkaline pH conditions, most of the mercury will undoubtedly be found in the HLW feed streams.

Mercury is relatively insoluble in borosilicate glass and tends to quantitatively volatilize at the operating temperature of the melter (Volf 1984). Volatility losses from the melter are believed to be affected by cold-cap coverage, plenum temperature, and feed/glass composition. Melter studies have indicated that as much as 50 percent of the mercury in the melter feed could volatilize in the melter (Table 4-1).

The Savannah River (SRS) plant reportedly used about 45 times as much mercury to catalyze the dissolution of aluminum-clad fuel elements. The DWPF flowsheet also shows a mercury feed rate of 19 kg/day to the mercury removal pretreatment step and 3 kg/day to the melter (Higley 1988). Process emissions from the SRS plant are controlled by venting mercury vapor from the processing tanks through the vessel vent header system condenser (formic acid vent condenser) and melter offgases through the offgas vent condenser. The overall emissions level is based on the saturation limit of mercury in the noncondensable gases emitted from the condensers. The current design is based on an exit temperature of 10 °C, while release estimates are based on an operating temperature of 15 °C. Mercury release rates also are based in part on the volume of purge air required to control the hydrogen concentration in the chemical processing cell. Purge air requirements are predicated on maintaining the hydrogen concentration below 25 percent of the lower explosive limit. When the plant reaches full production, mercury emissions are estimated to be 0.155 kg/day from the formic acid vent condenser and 0.136 kg/day from the melter offgas condenser.

At the DWPF facility, it is anticipated that 85 percent of the mercury will be removed in the feed preparation process (via formic acid reduction) and 15 percent will be routed to the melter. If this distribution also applies to the Phase I immobilization facility, the melter feed could vary from 6.3 to 49 p/M mercury (for Tanks 241-C-106 and 241-AY-102 waste). On this basis, mercury emissions from the Phase I melter (rated at 1.0 MT/day of glass @ 35 percent WOL) may range from 0.0022 to 0.0172 kg/day, with an average value of 0.0095 kg/day. Actual emissions could be substantially lower depending on the volume of purge air and operating temperature of the melter offgas condenser.

If the mercury decontamination factor across the melter is much less than 2.0, it may not be possible to recycle mercury to extinction in the HLW vitrification plant. The estimated distribution of mercury is addressed in the current flowsheet (Orme 1995). Although the removal of mercury from the feed stream may not be necessary, the presence of even a small amount of mercury may require the use of DWPF-specified materials in the offgas system (Hastelloy C-22 for feed preparation and offgas handling equipment) and the segregation and purification of mercury from scrubber wastes for offsite disposal.

6.4.2 Cadmium

Cadmium oxide (CdO) volatility is a concern because of the potential accumulation of cadmium-rich deposits in the melter offgas line. Such deposits were found to be minimal following the completion of the LFCM-8 melter run (Perez et al. 1993). At the entrance of

the offgas line jumper, a shallow 0.01-cm thick deposit coated the inside of the pipe. The maximum volume of the deposit was estimated to be 0.00056 m³. An analysis of this deposit indicates cadmium was not enriched relative to other components in the feed. The CdO release rate from the melter was determined to be 0.65 g/h. This corresponds to a melter decontamination factor of approximately 350, which is similar to decontamination factors previously established for aerosols of sodium and potassium (Goles and Nakaoka 1990). Less than 3 percent of the CdO exiting the melter was found to have deposited in the offgas line. Preliminary glass testing indicates that CdO is soluble up to 2.5 wt% in glass at Fe⁺²/Fe⁺³ ratio of 0.18 (Watrous 1987). Based on these estimates and the projected CdO composition of Tanks 241-AZ-101 and 241-AZ-102 wastes (1.80 and 3.54 wt%, respectively), offgas line deposits (of CdO) are unlikely to occur if Tanks 241-AZ-101 and 241-AZ-102 wastes are blended (2.84 wt% CdO). On the other hand, long term problems could arise if Tank 241-AZ-102 waste is processed separately and mechanical systems are not available to remove potential deposits from the melter offgas line. While CdO deposits may be a contributing factor, sodium metaborate and chloride salt deposits are likely to be a much more significant concern with respect to the melter offgas line.

6.4.3 Chloride, Fluoride, and Sulfate

Chloride and fluoride are expected to concentrate in the melter feed due to internal recycle of scrubber waste solutions in the vitrification plant. Preliminary estimates have been made as to the likely concentration of chloride and fluoride based on the HWVP material balance for NCAW (Tanks 241-AZ-101/241-AZ-102 waste) (Baker 1991). Two separate versions of the flowsheet were considered, one that assumes maximum recycle and the other that allows for purging a large fraction of the soluble components. The concentration of chloride increased by a factor of two in the recycle flowsheet to as much as 1,700 p/M in the feed preparation tank, while the concentration of fluoride increased by a factor of 1.7. This chloride level is at the upper end of materials selection test data where some accelerated corrosion (> 5 mils/year) has been observed (Elmore and Jensen 1990). More frequent plugging of the melter offgas line is also possible due to the increased rate of salt deposition. In comparison, the feed preparation tank chloride level in the purge flowsheet was about 630 p/M.

Most of the chloride, fluoride, and sulfate should be separated from the HLW stream during the sludge wash or enhanced sludge-washing processes. Estimated sludge-washing factors are on the order of 25 to 70 percent for each of these components based on characterization analysis of the core samples. However, these estimates are highly uncertain because the water washed solids were not analyzed for Cl, F, and SO₄. The total chloride inventory in the Phase I waste could be as little as 723 kg (out of 926,000 kg of unwashed sludge). Based on sludge washing studies, it is conservatively assumed that most of the chloride associated with silver in the Tank 241-AY-102 waste, will remain in the water washed sludge that will be delivered to the Phase I HLW plant. The amount of fluoride and sulfate in the water washed sludge from all Phase I wastes is likely to be low but cannot be determined from the available core sample data.

Salt deposits are commonly formed in the melter offgas line of LFCMs. These deposits usually consist of alkali-rich chlorides, sulfates, borates, and fluorides with entrained Fe_2O_3 , spinel, and frit particles (Hutson 1994). During the IDMS melter test where these deposits were observed, the melter plenum was maintained at a temperature of 800 to 900 °C and the offgas line at 300 to 350 °C. Ruthenium also was found in the IDMS offgas line deposits. Under very oxidizing conditions, RuO_2 can be oxidized to RuO_4 , a toxic gas. RuO_4 is very unstable and will probably decompose to RuO_2 in the melter offgas line. However, most ruthenium was found to be associated with a spinel phase, which suggests that ruthenium-rich deposits were caused by entrainment rather than by volatilization of ruthenium. Table 6-25 shows the chemical composition of the offgas deposits found during the IDMS tests with HWVP feeds (NCAW).

Table 6-25. Chemical Composition of Offgas Deposits.

Component	After HWVP1 (wt%)	After HWVP2 (wt%)
Na	--	14.97
Cs	--	0.92
Zn	0.13	0.13
Cd	0.58	0.95
F	0	0.01
Cl	8.93	9.86
SO_4	12.42	17.35

HWVP = Hanford Waste Vitrification Plant

In order to support material selection for HWVP, corrosion tests were performed using conservative (aggressive) process stream simulants and conditions. Corrosion limits of less than 125 $\mu\text{/year}$ were set accordingly for tanks and process lines with no allowance for localized stress corrosion cracking, pitting, or crevice corrosion phenomenon. Certain materials were chosen for testing in a high-chloride and -fluoride environment, including 304L SS, Carpenter 20 Cb-3, Hastelloy¹ C-276, Hastelloy C-22, Hastelloy G-3, Titanium Grade 7, Titanium Beta-C, and Allcorr (Elmore and Jensen 1990). These materials were tested in the presence of simulated offgas condensates with 80,000 p/M of chloride and 9,000 p/M of fluoride over a pH range of 1.0 to 3.0 and in applicable slurries with 20,000 p/M and 2,300 p/M of chloride and fluoride, respectively. Depending on pH, several materials were identified as candidate materials for use in a high-halide environment. Hastelloy C-22 exhibited the lowest corrosion rates, generally less than 125 $\mu\text{/year}$. The

¹Hastelloy is a trademark of Stellite Rod Division, Stoddy Deloro Stellite, Incorporated.

corrosion rates did not change appreciably as a function of chloride and fluoride concentration.

The bounding case for chloride appears to be a blend of tanks 241-C-106 and 241-AY-102 waste with an average chloride concentration of 705 p/M in the melter (assuming that all chloride in the supernate wash stays in the water washed sludge). The equivalent concentration in the recycle flowsheet could be as high as 1,400 p/M in the melter feed tank. This concentration is bounded by the off-gas condensate chloride concentration used in the materials corrosion test from which Hastelloy C-22 was identified as the preferred material of construction for the melter feed tank (Elmore and Jensen 1990). The worst case scenario for fluoride is again Tank 241-AY-102 waste, with 700 p/M of fluoride in the melter and 1,400 p/M in the melter feed tank based on the recycle flowsheet. For sulfate, the bounding condition is represented by Tank 241-AZ-101 waste, where the maximum sulfate concentration could be 115 p/M in the melter (@ 36 percent WOL). The only data from which a comparison can be made is the WVDP melter feed where the Inconel-690 electrodes were not affected by 2,500 p/M of sulfate in the glass. Average flowsheet values and offgas scrubber purge stream concentrations are shown in the TWRS flowsheet (Orme 1995).

6.4.4 Nitrogen Oxides

Nitrogen oxide emissions could be on the order of 12 to 18 MT/year when processing NCAW feed (from Tanks 241-AZ-101/241-AZ-102) through the vitrification process (at a rate of 1.0 MT/day of glass @ 38 percent WOL) (Stegen 1991). Nitric oxide emissions and NO_x treatment processes are identified in the TWRS flowsheet (Orme 1995). If nitrogen oxide emissions exceed 40 MT/year, air quality regulations could require use of best available control technology (BACT), which would be determined on a case-by-case basis. Selective catalytic reduction (SCR) of nitrogen oxides with ammonia may be the preferred process for BACT control. The incremental cost of this system would be about \$3 million for the HWVP, with a projected annual cost of \$950,000 for maintenance and operation.

There are only a few examples where nitrogen balances have been determined across the melter. In two small-scale tests at SRS, about 30 to 50 percent of the nitrate was converted to nitrogen oxide in the melter (Crow and Sabatino 1984, Crow 1985). A 50 percent fractional conversion was obtained during a scale-glass melter run at SRS (Crow 1987). All the SRS tests were conducted with plenum heaters in the melters and with relatively high-volatile organic concentrations in the feed. During the PSCM-23 melter test, PNL reported a 75 percent fractional conversion of nitrate to nitrogen oxide without plenum heaters in the melter. On this basis, it appears that higher fractional conversions (and a general increase in nitrogen oxide emissions) may be caused by lower plenum temperatures in the melter and by a lower concentration of volatile organics in the feed.

The dissociation of nitrogen peroxide into nitric oxide and oxygen starts to occur in the temperature range of 150 °C to 620 °C, and at temperatures above 700 °C nitric oxide will

partially decompose into nitrogen and oxygen. About 95 to 99 percent of the nitric oxide can be destroyed in a high-temperature reducing flame (U.S. Patents 3,563,697, 3,232,713, and 2,673,141).

The melter feed preparation process appears to be the main source of nitrogen oxides. Nitrogen oxides are produced when formic acid is added to the feed to control redox conditions in the melter. Under these conditions, nitrites will be almost quantitatively reduced to nitrogen oxides in the acidified feed. Formic acid also has been known to produce appreciable amounts of hydrogen and ammonia during the feed preparation process. The AX, tank farm, and DST blends are expected to be the main nitrogen oxide producers based on the projected nitrate and nitrite concentrations in these wastes.

6.4.5 Criticality Control

Table 6-26 provides an estimate of the plutonium inventories in Tanks 241-AZ-101, 241-AZ-102, 241-C-106, and 241-AY-102 [Powell 1996]).

Table 6-26. Plutonium Inventories in Tanks 241-AZ-101, 241-AZ-102, 241-C-106, and 241-AY-102.

Tank	Plutonium, kg
241-AZ-101	19.25
241-AZ-102	27.19
241-C-106	97.5
241-AY-102	8.64

The source of the inventory estimate for 241-C-106 is Rogers (1994). The source of the other estimates is Tusler (1995). These estimates, which are based on the highest plutonium value in any tank sample, are considered the high estimates for the tanks.

Table 6-27 shows the maximum plutonium oxide (PuO_2) concentration in the NCAW melter feed (based on undiluted waste composition).

Table 6-27. Maximum Plutonium Oxide Concentration in Phase I Melter Feeds.

Tank	Total oxides (kg)*	PuO ₂ (kg)	PuO ₂ (wt %)
241-AZ-101	109,214	21.82	0.02
241-AZ-102	172,989	30.83	0.02
241-C-106	485,519	110.55	0.02
241-AY-102	31,646	9.80	0.03

*Water washed oxides.

At SRS, the solubility of PuO₂ was determined to be 4 wt% in Frit 21 and 7.0 wt% in Frit 411 (Plodinec 1979, Soper 1982). Both frits are essentially borosilicate glasses with different concentrations of Na₂O, Li₂O, and TiO₂. Because the solubility of PuO₂ in borosilicate glass appears to be at least 140 times the limiting concentration in NCAW waste, the precipitation of PuO₂ as a separate phase is not anticipated. The possibility of a criticality incident in the DWPF melter due to the accumulation of PuO₂ or plutonium metal has been examined at SRS (Soper 1982). The solubility of plutonium was found to greatly exceed the amount present in SRS waste. A thermodynamic analysis of PuO₂ reduction shows that it will be one of the last oxides in the glass to reduce to metal. In particular, iron, nickel, and manganese oxides will be reduced to metal before PuO₂ is reduced. Large amounts of iron and other metals in the melter would cause severe operating problems and shorting of the electrodes before any significant occurrence of plutonium accumulation.

Criticality control in the melter should be based on the possible accumulation of a plutonium-rich secondary phase at the bottom of the melter. Because the British have observed that cerium tends to form insoluble precipitates with actinides (plutonium and neptunium) in molten glass, the possibility of cerium, uranium, or some other carrier-induced precipitation of plutonium needs to be examined. If glass formulation testing shows that plutonium-rich crystals will not form or precipitate in the melter, criticality control should be based on dilution of plutonium in the glass such that K_{eff} would be much less than 0.95.

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7.0 RESULTS AND CONCLUSIONS

The primary purpose of this study is to assess the possible range of feed compositions and the effect of these compositions on the Phase I HLW immobilization process. The Phase I wastes consist of water washed (WW) and caustic washed (CW) sludges from Tanks 241-AZ-101, 241-AZ-102, 241-C-106, and 241-AY-102. For glass formulation purposes, several blends were also considered, including a nominal blend of these wastes derived from 80 percent of Tank 241-AZ-101, 80 percent of Tank 241-AZ-102, 37 percent of Tank 241-C-106, and 15 percent of Tank 241-AY-102 wastes. Estimates of the feed composition were derived from the Tank Characterization Reports for Tanks 241-AZ-101, 241-AZ-102, and 241-AY-102, except for the sludge and aluminum inventories in Tank 241-AZ-101, and from the Data Transmittal package for Tank 241-C-106 (Hodgson 1995, Ryan 1995a, 1995b, Weiss 1987). The current inventory estimates for Tank 241-AZ-101 are based on the physical and chemical properties of the cores and average sludge level when the samples were taken in May 1989. At least one full-depth core sample has been taken from each of the Phase I tanks. Sludge washing studies were performed on three radioactive cores from Tanks 241-AZ-101 and 241-AZ-102 to characterize the composition of the washed solids. These results were used to estimate the composition of the water washed Phase I HLW feeds.

In this study, several different Phase I staging (retrieval, blending, and pretreatment) scenarios were used to generate example feed compositions for glass formulation, testing and glass sensitivity analysis. It should be recognized that these scenarios will not necessarily match the final process that is selected for Phase I.

Glass models and data from laboratory glass studies were used to estimate the achievable WOL and corresponding glass volume for various Phase I feeds. This study also takes into consideration possible variations in the feed composition due to uncertainties in the waste composition, blending or pretreatment processes that will be applied to these wastes. Glass property data have been obtained and evaluated for all Phase I glasses of primary interest. This includes glass property data correlations developed by PNNL that have been incorporated into computer codes (models) to facilitate calculation of estimated glass properties. These models were extensively used in the current evaluation. Key issues related to feed processability, feed composition uncertainty, and immobilization process technology are identified for future consideration in other tank waste disposal program activities.

7.1 GLASS COMPOSITION

Melter feed compositions are expected to vary due to the range of compositions in the tanks, incomplete recovery and mixing of wastes, and batch composition differences depending on the waste consolidation plan. The corresponding glass composition will, therefore, vary for different waste types around a nominal glass composition for each waste. Changes in glass composition affect melt viscosity, electrical conductivity, and phase behavior. These properties determine the processability of the glass in a joule-heated melter. Glass composition also affects chemical durability and the related acceptability of the glass for the geologic repository. The optimum glass composition has to be determined for each waste.

The reference waste form in this study is borosilicate glass. Borosilicate glass has been recommended based on its ability to immobilize a wide range of wastes, insensitivity to radiation damage, and general properties such as thermal stability, chemical durability, and processability. The acceptable range of glass composition is normally defined by the range of compositions previously evaluated for the HWVP Project (the CVS range of compositions for which the statistically derived glass property models are considered valid).

Waste loading is defined as the weight percent of waste (oxides) in the HLW glass. This value is extremely important as it determines the required production rate of the melter and the total amount of glass produced during the Phase I HLW program.

7.2 MELTER PERFORMANCE SENSITIVITIES

Compounds or elements that require special treatment in the offgas may include tritium, mercury, ^{14}C , and ^{129}I . These elements will not be incorporated into glass at $1,150^\circ\text{C}$, except for a minor amount of mercury, and ^{129}I and must be captured in the offgas system and sent out as a secondary waste stream. Carbon-14 probably will be released to the atmosphere as long as *Clean Air Act Amendments of 1990* standards are met. Chloride, fluoride, technetium, cesium, cadmium, and ruthenium oxide are partially soluble in glass but require special consideration in the offgas treatment system for recycle.

Noble metal oxides and alloys of silver, rhodium, palladium, and RuO_2 are virtually insoluble in borosilicate glass. Previous tests and studies have indicated that accumulation of these metals can cause premature failure of the melter due to electrical short circuiting at concentrations as low as 0.1 wt% in glass. Designs developed by the Germans (KfK) have proven that accumulation of noble metals can be avoided by having a steeply sloped (75 degrees) bottom that allows precipitated noble metal particles to be drained out of a bottom drain. The Phase I water washed nominal blend contains 0.38 wt% noble metals (Ag_2O , Ru_2O_3 , Rh_2O_3). At 48 percent WOL in the glass, the noble metals concentration is 0.182 wt%. If Phase I feeds are processed separately, Tanks 241-AZ-101, 241-AZ-102, and 241-C-106 glasses are estimated to contain 0.17, 0.11, and 0.24 wt% noble metal oxides, respectively. On the basis of recent grab samples from Tank 241-C-106, the bounding noble

metal oxide concentration could be as high as 0.52 wt% in glass because of the amount of silver and ruthenium in this waste. This noble metals concentration is likely to cause premature failure of the melter electrodes and bottom refractory (based on the low melting temperature for silver [960 °C] and laboratory-scale tests performed with feeds containing from 0.33 to 0.46 wt% noble metals). An added concern is the high silver inventory in Tank 241-AY-102, with 1.85 wt% silver oxide (Ag_2O). If the noble metals concentration is determined to be too high, other wastes should be considered as potential feed stock for mitigating the high silver and ruthenium inventories in Tanks 241-C-106 and 241-AY-102. A bottom drain melter could also be used to address high silver inventories in the Phase I feed tanks.

7.3 MELTER TECHNOLOGY

Additional information has recently become available on performance of the high-temperature melter (HTM) and high-temperature glasses. Tests with a refractory-lined HTM indicate that the melt-rate capacity (at 1,350 °C) is only about 10 percent higher than the capacity of the low-temperature melter (LTM) (at 1,150 °C). Dissolution of refractory components into the glass increases aluminum and chrome levels and thus reduces the previously expected waste oxide loading advantage of the LTM (if corrosion source term components are included in calculating the liquidus temperature limits for glass formulation). Rapid corrosion of the melter electrodes also indicates that development of the HTM is likely to be more expensive than originally anticipated due to the need to identify and test more exotic (and expensive) electrode materials.

For high-nickel, -chromium, or -iron feeds, spinel precipitation and sludge accumulation problems are likely to occur in the melter. The precipitation and settling characteristics of spinel need to be assessed under different operating conditions. Viscosity and electrical conductivity measurements are needed to determine whether spinel phase enriched sludge deposits are primarily a mechanical or electrical problem in the melter. Glass melts containing from 2 to 10 vol% crystals are known to have approximately Newtonian viscoelastic behavior (Plodinec 1986).

Glass property data also indicate that precipitation of sludge forming solids (spinel phases) limits the achievable waste oxide loading for Tanks 241-AZ-101/241-AZ-102 wastes. If a practical sludge removal system were developed for the LTM (or alternative melter), it would be feasible to process glasses that contain a small fraction of undissolved solids (precipitated crystals). Several different approaches may be potentially promising, including: (1) a cone-shaped melter bottom, bottom electrode and drain system, (2) a detachable side entering drain with features to induce flow, (3) a ceramic stir or bubbler system to mobilize the solids and the injection of low viscosity glass to purge the sludge, or (4) a top entering vacuum removal system, such as the one developed by PNNL for West Valley (evacuated canister design) (Barnes 1989).

Glass formulation results indicate the WOL of Tank 241-AZ-101/241-AZ-102 wastes could be increased 30 to 40 percent if the solids precipitation limits (liquidus temperature constraints) were relaxed. For example, optimum formulations for the HTM contain as much as 50 wt% NCAW waste (and 1.5 to 4.6 vol% crystals because of refractory corrosion). Similar glasses can be formulated for the LTM that contain from 33 wt% NCAW (with no crystallinity), to 40 or 45 wt% NCAW (with 0.4 to 0.9 vol% crystals). If a practical sludge removal system were developed, LTM glasses could be formulated beyond the liquidus temperature limit to possibly match the waste loading capabilities of the HTM, significantly reducing the predicted life-cycle costs for the LTM. If the liquidus temperature limit is raised from 1,050 to 1,100 °C, for example, approximately 53 fewer canisters (10.7 percent less glass) should be produced from the nominal blend feed.

Based on SSHTM and RSM-scale melter tests, the melt-rate capacity of the melter could probably be improved by formulating lower viscosity glasses for the melter. These glasses, however, are likely to increase refractory corrosion rates in the melter and could lead to premature failure of the refractory envelope.

7.4 PHASE I GLASS PROPERTIES

Based on current models, iron and aluminum are the main components that control the volume and composition of the Phase I glasses. Liquidus temperature and precipitation of nepheline (NaAlSiO_4) in the canister glass are the limiting properties of interest for most of these glasses. Liquidus temperature is important because this is the temperature where insoluble crystals will precipitate from the glass and may clog the pour spout or possibly short the electrodes in the melter (if the sludge phases are electrically conductive). Precipitation of nepheline during CCC is important because this is likely to produce an unacceptable HLW glass due to a substantial increase in normalized boron PCT release (due to a diminished amount of aluminum in the glass). Certain data were lacking with respect to the glass composition region of interest for the Phase I feeds. PNNL provided glass formulation support by experimentally evaluating properties of glasses in this composition region. Unpublished glass property data were also obtained from Westinghouse Savannah River Company. These data were used to develop improved data correlations and calculational models for the Phase I wastes. Glass work produced significant results in the following areas:

- Significant component interaction effects were noted in several Phase I glasses. The most important interaction is the effect of aluminum on the solubility of iron (reducing aluminum increases the solubility of iron). Aluminum and iron have a dominant effect on the liquidus temperature of the Phase I glasses, while aluminum, silica, and sodium control the nepheline forming properties of the glass.
- Better liquidus temperature models are being developed from the current database and a new set of glasses are being formulated around a spinel-forming baseline

composition, with targeted variations in Al_2O_3 , B_2O_3 , Cr_2O_3 , Fe_2O_3 , Li_2O , MgO , MnO , Na_2O , and SiO_2 to measure the effect of each component on the liquidus temperature of spinel. Spinel structures are represented by the formulated $\text{M}^{\text{II}}\text{M}_2^{\text{III}}\text{O}_4$, where M^{II} consists of di-valent Fe, Mg, or Mn and M^{III} tri-valent Fe, Cr, or Al.

- Precipitation of nepheline appears to be a significant issue for certain feeds. Significant progress was made in FY 1996 in determining the composition ranges where nepheline precipitation could be a problem and estimating the possible durability effects of nepheline precipitation.
- Glass property models may be used to formulate and estimate the volume of HLW glasses. Based on these models and glass property data, the maximum waste oxide loading for the Phase I glasses varies from 29.78 to 30.65 wt % on an adjusted feed oxide basis. This indicates there is a 17 percent contingency or operating margin between the maximum theoretical waste loading and the minimum 25 percent adjusted feed oxide basis specified in the RFP. The 25 percent AFO appears to be conservative and an achievable value for Phase I.

The primary conclusions from these findings are as follows:

- The incentive for reducing the amount of aluminum in the Phase I waste (with caustic washing) is larger than previously thought.
- The iron solubility limit, as determined by liquidus temperature constraints, is lower than previously thought for many feeds, especially those with significant aluminum content. High concentrations of chromium and manganese also have an adverse effect on allowable iron compositions because of liquidus temperature effects.
- The nepheline region can be avoided by defining a minimum limit for SiO_2 such that glasses with a normalized SiO_2 content ≥ 0.62 should not be susceptible to nepheline precipitation in the canister. The defined limit for nepheline is important for certain wastes, such as Tank 241-C-106 waste, where the WOL limits are likely to be determined by the nepheline forming characteristics of the glass.
- Glass property data and models should be considered in developing future baseline glass quantity estimates. Recent glass data identifies some important component interaction effects that were not well represented in simplified single component and multi-component constraint limits. A primary example is the effect of aluminum on iron solubility in glass. This interaction generally results in a larger glass volume and more incentive for caustic washing as compared to glass estimates derived from single component or multi-component limits.

Three core samples of radioactive waste were recovered from the Phase I Tanks 241-AZ-101 and 241-AZ-102 in 1991. Small portions were used in laboratory-scale feed preparation and vitrification tests to produce representative samples of radioactive glass. The settling characteristics of the simulants did not match the behavior of the core samples in that the core samples settled much more quickly and achieved a much higher solids density than the simulants. The radioactive and simulant glasses proved to be highly durable. Radioactive glasses were similar but slightly more leach resistant than comparable simulant glasses.

7.5 GLASS SOLUBILITY LIMITS

Because spinel phases have little effect on the properties of the glass, a chromium oxide limit is mainly needed to avoid operability problems in the melter and to minimize the possible precipitation and accumulation of spinel (unless a bottom drain melter is used). The Cr_2O_3 solubility limit for LLW glass varies from 0.48 to 1.04 percent, while the limit for high temperature glasses with a substantial amount of Fe_2O_3 varies from 0.6 to 0.93 percent. These results are important because they show that the solubility of Cr_2O_3 (as one of several spinel forming components in the waste) is only marginally increased by the temperature and composition of borosilicate and aluminum silicate glasses (thus reducing the potential advantage of the HTM).

Phosphate tends to form precipitates with calcium, rare earths, and lithium. Molten salt or scum layers may form from low-melting temperature phosphates and interfere with the operation and melt-rate capacity of the melter. A limit of 3.0 wt% phosphate should be acceptable because the Phase I glasses are projected to have less than 1.0 wt% phosphate and 0.85 wt% rare earths and should not be susceptible to phosphate induced scum or phase separation in the melter.

Although halides can form molten salts that accumulate on the glass surface, only minimal quantities of such salts have ever been observed. Presently, there is no generally applicable solubility limit for chloride. For Hanford Site LLW, a limit of 0.3 g Cl/100 g of waste oxide in the melter feed is being recommended for borosilicate glass based on the corrosion properties of chloride (in the melter offgas system) and the effect on cesium volatility. Halides are expected to distribute almost quantitatively to LLW.

The noble metals solubility limit is currently less than 0.25 wt% $\text{Ru}_2\text{O}_3 + \text{PdO} + \text{Rh}_2\text{O}_3$ in glass (Kalia 1992a). This limit is well above the solubility limits derived from other studies that suggest flat-bottom melters may be susceptible to premature electrical failure at noble metals concentrations as low as 0.1 wt% in glass.

7.6 WASTE OXIDE LOADING AND GLASS VOLUME ESTIMATES

The glass formulation models were used to test the bounding composition limits for each of the Phase I glasses. The main purpose of this effort is to determine the bounding conditions for each waste and the possible effects of a better retrieval sequence, blending strategy or pretreatment process on the number of glass canisters produced. Sensitivities were evaluated by increasing the allowable liquidus temperature to 1,100 °C, or by assuming that 25, 40, 50, and 75 percent of the Al_2O_3 could be removed (by caustic washing) and the glass reformulated to the maximum WOL. The glass models include the Fulcher first-order viscosity model (Section 5.3), spinel second-order liquidus temperature model (Section 5.5), second-order PCT B and Na release models and the nepheline model (Section 5.6). The results are summarized in Table 7-1.

Table 7-1. Glass Volume Estimates for Phase I Experimental Glasses.

Waste	Base glass	Base glass, cans.	T _L limit @ 1,100 °C cans.	25% Al ₂ O ₃ reduct, cans	40% Al ₂ O ₃ reduct, cans	50% Al ₂ O ₃ reduct, cans	75% Al ₂ O ₃ reduct, cans
WW nom. blend	Nom-3	495	442	478	-	455	432 ^a
CW nom. blend	Nomc-1	417	384	411	-	400	397 ^{a,b}
WW AZ blend	AZ-3	453	394	430	421 (1)	-	-
WW 241-C-106A	Reform. C106A-4	436	N/A	399	-	356	316
WW 241-C-106B	Reform. C106B-1	467	N/A	415	-	359	270 ^c

WW = Water washed.

CW = Caustic washed.

Nom. blend = Nominal blend (80 percent 241-AZ-101, 80 percent 241-AZ-102, 37 percent 241-C-106, 15 percent 241-AY-102).

AZ blend = Tank 241-AZ-101/241-AZ-102 blend.

241-C-106A = Based on Weiss core sample analysis of Tank 241-C-106 waste.

241-C-106B = Based on Hara/McCown core sample analysis of Tank 241-C-106 waste.

cans = Number of DWPF canisters (1,650 kg of glass per canister).

N/A = Not applicable.

^aGlass limited by PCT B release.

^bGlass limited by T_L limit (@ 1,050 °C).

^cGlass limited by nepheline region.

The glass composition boundaries are typically established by the liquidus temperature or nepheline composition limits for each glass. Experimental glasses were formulated by Pacific Northwest National Laboratory. In several instances, the experimental glasses were not formulated to the edge of the liquidus temperature or nepheline composition region and so these glass were reformulated with the glass models to optimize the potential WOL for each composition. The glass models predict that 53 fewer canisters will be produced if the liquidus temperature limit is raised from 1,050 to 1,100 °C for the water washed nominal blend. Approximately 40 to 78 canisters also could be saved if caustic washing processes are used to remove 50 percent of the Al_2O_3 from nominal blend feed (the CW nominal blend also contains about 50 percent of the of Al_2O_3 in the nominal blend feed). The caustic washing results for the nominal blend are slightly different because the glass models are non-linear and the output is often affected by the initial composition of the waste (either water or caustic washed waste). For the AZ blend, the maximum WOL was reached at 40 percent Al_2O_3 removal based on the durability properties of the glass as measured by PCT boron release. By removing 40 percent of the Al_2O_3 , approximately 32 canisters of glass could be saved if the AZ blend is caustic washed.

The glass models and glass property results were used to produce the following estimates for the water washed (base case) and caustic washed Phase I wastes (Tanks 241-AZ-101, 241-AZ-102, and 241-C-106 wastes and various blends of these wastes).

Table 7-2. Glass Volume Estimates for Unblended Phase I Wastes.

Phase I tanks	Water washed waste	Waste inventory (MT) ^a	Base glass WOL, wt%	Base glass AFO, wt%	Base ^b glass (limit)	Cans w/50% Al_2O_3 removed (limit)	Cans w/75% Al_2O_3 removed (limit)
241-AZ-101	241-AZ-101	109.46	36.08	30.26	183 (T_L)	154 (T_L)	145 (T_L)
241-AZ-102	241-AZ-102	172.99	34.40	29.28	305 (T_L)	265 (T_L)	268 (T_L)
241-C-106	241-C-106A	485.52	67.45	30.65	436 (neph.)	356 (T_L /neph)	316 (low vis)
Total		767.97			924	775	729

^a100 percent retrieval efficiency.

^bDWPF canisters of glass limited by T_L liquidus temperature or nepheline.

Phase I blend compositions

Nominal blend: $0.8 * 241-AZ-101 + 0.8 * 241-AZ-102 + 0.37 * 241-C-106 + 0.15 * 241-AY-102$

AZ blend: $241-AZ-101 + 241-AZ-102$

AYC blend: $241-C-106 + 241-AY-102$

AYC-A blend: $241-AY-102 + 0.1625 * 241-C-106$

AYC-B blend: $241-AY-102 + 0.4875 * 241-C-106$

AZC-A blend: $AZ Blend + 0.4875 * 241-C-106$

AZC-B blend: $AZ Blend + 0.1625 * 241-C-106$

Table 7-3. Glass Volume Estimates for Various Phase I Blends.

Phase I blends	Water washed waste	Waste inventory (MT)	Cans ^a of base glass (limit) ^b	Cans w/50% Al ₂ O ₃ removed (limit) ^b	Cans w/75% Al ₂ O ₃ removed (limit) ^b
Nominal	Nominal	396.3	495 (T _L)	455 (T _L)	432 (T _L)
AZ Blend	AZ Blend	282.52	453 (neph.)	421 (dur. w/40% Al ₂ O ₃)	N/A
AYC	241-AY-102/ 241-C-106A	514.55	554 (neph.)	397 (neph.)	343 (T _L)
AYC-A	241-AY-102/ 241-C-106A	116.23	139 (neph.)	99 (T _L)	96 (T _L)
AYC-B	241-AY-102/ 241-C-106A	281.36	333 (neph.)	220 (neph.)	211 (durable)
AZC-A	241-AZ-101/ 241-AZ-102/ 241-C-106A	479.83	600 (neph.)	477 (T _L)	461 (T _L)
AZC-B	241-AZ-101/ 241-AZ-102/ 241-C-106A	315.14	450 (neph.)	372 (T _L)	365 (T _L)

^aDWPF canisters of glass.

^bGlass formulation limit (T_L liquidus temperature limit, nepheline composition limit, durability limit).

If caustic washing processes are employed, the glass models indicate 80 canisters could be saved by removing 50 percent of the Al₂O₃ from Tank 241-C-106 waste and 120 canisters if 75 percent of the Al₂O₃ is removed from this waste. Caustic washing processes appear to be less desirable for Tanks 241-AZ-101 and 241-AZ-102 wastes, producing 29 and 40 fewer canisters, respectively. These estimates are based on the physical and chemical properties of the glass. Caustic washing studies will be necessary to determine whether it is possible to remove this much aluminum from the Phase I wastes. Also, it is unclear whether the TWRS safety basis could be modified in time to allow enough tank space for a caustic washing option for these wastes.

If Tanks 241-AZ-101/241-AZ-102 wastes are caustic washed, approximately 419 canisters will be produced from the residual waste (Table 7-2), compared to 421 canisters from the caustic washed AZ blend (Table 7-3). The total amount of glass

derived from Tanks 241-AZ-101 and 241-AZ-102 waste (488 canisters base case) is slightly more than predicted for the AZ blend (453 canisters). Again, this is an instance where the nonlinear glass models have produced slightly different results depending on the initial composition of the waste.

Glass property data indicate the maximum waste oxide loading the nominal blend feed is 48.5 wt% total oxides or 30.9 wt% on an adjusted feed oxide (AFO) basis. The AFO basis is defined as the weight of total non-volatile feed oxides excluding sodium and silica on an oxide basis divided by weight of the product glass. If the Phase I wastes are processed separately, the average waste oxide loading is 30.1 wt% AFO (mass weighted average). This indicates that little benefit will be gained from blending in terms of reducing the overall volume of glass. However, there may be other benefits from blending, including reducing the concentration of problem components (such as silver) and reducing the number of glasses and process conditions that need to be qualified.

Current estimates do not include any allowance for operating margin contingency (to compensate for uncertainties in the waste or glass composition analysis). If it is assumed that production glasses are formulated to 85 percent of the maximum waste loading, the average waste loading for the tank by tank and nominal blend feed are 25.6 and 26.3 percent adjusted feed oxides, respectively. For the tank by tank feed, this results in 957,000 kg of glass (580 canisters) for the minimum Phase I feed of 245,000 kg AFO or 1,816,000 kg of glass (1,100 canisters) for the maximum feed of 465,000 kg AFO.

Sensitivity analyses were also performed to evaluate the effects of feed composition and more aggressive caustic washing processes for these wastes. The purposes of this analysis is to provide feedback to the pretreatment and waste retrieval functions, and to identify high priority issues to be addressed in any future waste characterization activities.

The most important components from a glass making perspective are Al_2O_3 and Fe_2O_3 . Because glass volume predictions are important to overall project cost, a sensitivity study was performed to assess the likely impact of varying the amount of Al_2O_3 and Fe_2O_3 in the Tanks 241-AZ-101 and 241-C-106 wastes. For Tank 241-AZ-101 waste, the Al_2O_3 to Fe_2O_3 ratio of 0.74 is an important characteristic of the glass. If this ratio is less than 0.74, the glass volume is controlled by the liquidus temperature characteristics of the glass, while the nepheline forming properties control the volume of glass above this ratio. The results are summarized in Table 7-4.

Table 7-4 shows the effect of varying the Al_2O_3 inventory in Tank 241-AZ-101 based on three different glass formulation constraints. The CVS single component glass limits represent the bounding compositions used in the CVS glass formulation study (15 wt% Fe_2O_3 and 17 wt% Al_2O_3), while the liquidus temperature and nepheline limits represent the glass property limits for avoiding sludge accumulation problems in the melter and for meeting acceptable durability conditions at the canister centerline, respectively. Other properties such as viscosity and durability as measured by boron and sodium release are also important, but only rarely become the limiting constraints for glass formulation. Glass volume predictions

are based on the minimum volume of glass consistent with the liquidus temperature or nepheline limits. In the region from -75 to 0 percent change in the nominal Al_2O_3 inventory, the liquidus temperature limit controls the volume of glass, while the nepheline limit controls the composition of the glass from 0 to 100 percent increase in the nominal Al_2O_3 inventory. The CVS single component glass limits tend to be overly optimistic in relation to the amount of glass predicted by the glass models from this waste. Based on the results in Table 7-4, the CVS limits tend to underpredict the amount of glass by 30 to 60 percent in the high Al_2O_3 range and 5 to 30 percent in the lower Al_2O_3 (and higher Fe_2O_3) concentration range.

Table 7-4. Glass Volume as a Function of the Amount of Al_2O_3 in Tank 241-AZ-101 Waste.

Al_2O_3 , % change in mass	HLW glass model, cans	CVS glass limits, cans (limit)	Liq. temp. limit, cans	Neph. limit, cans	Al_2O_3 conc. in glass, ^a wt%	Fe_2O_3 conc. in glass, ^a wt%	Glass vol % change ^a
-75	145	139 (Fe)	145	136	2.64	14.4	-20.8
-50	154	139 (Fe)	154	150	4.97	13.54	-15.8
0	183	139 (Fe)	183	182	8.35	11.38	0
25	211	139 (Fe)	189	211	9.08	9.9	15.3
50	238	139 (Fe)	204	238	9.67	8.78	30.0
75	270	158 (Al)	219	270	9.95	7.75	47.5
100	286	180 (Al)	229	286	10.71	7.30	56.3

^aGlass composition based on the minimum volume of glass consistent with liquidus temperature and nepheline limits.

Tank 241-AZ-101 glass limits change from nepheline to liquidus temperature control at the baseline glass composition (8.35 wt% Al_2O_3 and 11.38 wt% Fe_2O_3). This transition is important because it shows the nepheline is the limiting property above the Al_2O_3 to Fe_2O_3 ratio of 0.74 in the glass (and Tank 241-AZ-101 waste), while liquidus temperature is the limiting property below this ratio. The overall glass volume increases 56 percent with a 100 percent increase in the amount of Al_2O_3 in the waste, but decreases only 21 percent with a 75 percent decrease of Al_2O_3 . Under nepheline control, the glass volume increases strongly with the amount of Al_2O_3 in the waste, while the volume decreases only about one-half as much under liquidus temperature control. Therefore, caustic washing process appears to be only about 50 percent effective (in terms of glass volume reduction) in the liquidus temperature region of the composition envelope (below the Al_2O_3 to Fe_2O_3 ratio of 0.74 for Tank 241-AZ-101 waste). If the mass of all components except Al_2O_3 or Fe_2O_3 is held constant, the volume of Tank 241-AZ-101 glass is projected to increase 4.3 and 5.6 percent with a 10 percent increase in the amount of Fe_2O_3 or Al_2O_3 , respectively, and decrease

0.65 and 2.8 percent with a decrease of 10 percent in the amount of Fe_2O_3 or Al_2O_3 in this waste. These figures can be used to estimate the cost of over-estimating or under-estimating the mass of these components (based on a repository disposal fee of \$356,000 per canister). These results simulate the effect of reducing Al_2O_3 by using the caustic washing process or inadvertently adding Al_2O_3 by precipitation from the liquid phase.

Glass model predictions for Tank 241-C-106 waste are summarized in Table 7-5. This table shows the effect of varying the Al_2O_3 inventory in Tank 241-C-106 based on the CVS single component glass, liquidus temperature and nepheline limits for HLW glass. Glass volume predictions are based on the minimum volume of glass consistent with the liquidus temperature and nepheline limits for this glass. The nepheline and liquidus temperature limits evidently converge in the range from -25 to -50 percent of the baseline Al_2O_3 inventory, while low viscosity is the limiting constraint at -75 percent Al_2O_3 . The nepheline forming characteristics of the glass control the formulation results in the range from 0 to 100 percent increase in nominal Al_2O_3 inventory. The CVS single component glass limits once again tend to underpredict the actual amount of glass by 25 to 35 percent in the range from -25 to +50 percent of the baseline Al_2O_3 inventory.

The crossover from nepheline to liquidus temperature/nepheline control occurs at -25 percent of the nominal Al_2O_3 inventory (9.39 wt% Al_2O_3 and 12.06 wt% Fe_2O_3 or Al_2O_3 to Fe_2O_3 ratio of 0.77). This transition is interesting because it occurs in the same composition region as the (nepheline to liquidus) transition in Tank 241-AZ-101 glasses. The glass volume increases nearly 47 percent with a 100 percent increase in the amount of Al_2O_3 in the waste, and decrease 18.3 percent with a 50 percent decrease of Al_2O_3 . Under nepheline control, the volume of glass increases sharply with the amount of Al_2O_3 in the waste, and decreases at a slightly smaller rate as the amount of Al_2O_3 is reduced in the waste. If the mass of all components except Al_2O_3 or Fe_2O_3 is held constant, the volume of Tank 241-C-106 glass should increase 4.3 and 4.6 percent with a 10 percent increase in the amount of Fe_2O_3 or Al_2O_3 , respectively, and decrease 1.1 and 3.6 percent with a decrease of 10 percent in the amount of Fe_2O_3 or Al_2O_3 in this waste. These figures can also be used to estimate the cost of over-estimating or under-estimating the mass of these components (based on a repository disposal fee of \$356,000 per canister).

Table 7-5. Glass Volume as a Function of the Amount of Al_2O_3 in Tank 241-C-106 Waste.

Al_2O_3 , % change in mass	HLW glass models, cans	CVS glass limits, cans (limit)	Liq. temp. limit, cans	Neph. limit, cans	Al_2O_3 conc. in glass, ^c wt%	Fe_2O_3 conc. in glass, ^c wt%	Glass vol % change ^c
-75	316 ^a	321 (Fe)	^b	^b	3.95	15.22	-27.5
-50	356	321 (Fe)	356	356	7.01	13.52	-18.35
-25	399	321 (Fe)	400	399	9.39	12.06	-8.49
0	436	321 (Fe)	419	436	11.45	11.03	0
25	496	367 (Al)	480	496	12.59	9.7	13.76
50	546	440 (Al)	506	546	13.73	8.82	25.23
75	590	514 (Al)	558	590	14.81	8.16	35.32
100	640	587 (Al)	621	640	15.59	7.51	46.79

^aGlass composition limited by low viscosity^bLiquidus temperature could not be increased to target level without adding Al_2O_3 .^cGlass composition based on the minimum volume of glass consistent with liquidus temperature and nepheline limits.

Glass models can also be used to identify the most important components in the waste and the analytical accuracy required for glass formulation purposes. These predictions are based on the minimum change (in waste concentration) needed to induce a 50 °C increase or decrease in the output of the second-order liquidus temperature model or a 5 percent change in the nepheline limit for Tank 241-AZ-101 and 241-C-106 glasses. The results are summarized in Table 7-6 for Tank 241-AZ-101 waste.

Table 7-6. Waste Concentration Change Needed to Induce a 50 °C Change in Liquidus Temperature (T_L) or 5 Percent Change in the Nepheline Limit for Tank 241-AZ-101 Glass.

Component	Source (frit/waste)	Component contribution to T_L , °C	Percent change needed for 50 °C increase or decrease in T_L	Percent change needed for 5% change in nepheline limit
SiO ₂	Frit 97.64/ Waste 2.36	-244.17	867	339
B ₂ O ₃	Frit 98.74/ Waste 1.26	115.24	3,400	
Na ₂ O	Frit 71.70/ Waste 28.30	114.45	154	85.9
Li ₂ O	Frit 99.59/ Waste 0.41	-23.25	5,240	
CaO	Waste 100	10.64	470	
MgO	Waste 100	-2.97	1,683	
Fe ₂ O ₃	Waste 100	358.96	13.9	
Al ₂ O ₃	Waste 100	369.93	13.5	49.6
ZrO ₂	Waste 100	73.16	68.3	
Cr ₂ O ₃	Waste 100	48.32	103.4	
MnO ₂	Waste 100	89.98	55.6	
NiO	Waste 100	15.95	313.5	
P ₂ O ₅	Waste 100	15.91	314.3	

These models indicate that only two components (Fe₂O₃ and Al₂O₃) need to be known with reasonable precision (to within 15 percent or their true values) to estimate the amount of glass likely to be produced from Tank 241-AZ-101 waste, and four components (SiO₂, Na₂O, Fe₂O₃, and Al₂O₃) to estimate the volume of glass from Tank 241-C-106 waste. Other components from a glass formulation perspective that need to be known (to within 100 percent of their true values) are Na₂O, MnO, and ZrO₂ for Tank 241-AZ-101 waste and Cr₂O₃, MnO, and CaO for Tank 241-C-106 waste. While it may be desirable to obtain as much information as possible, the volume and composition of most glasses can be reasonably

determined from Fe_2O_3 and Al_2O_3 composition data, together with some knowledge of Cr_2O_3 , MnO , noble metals (Ag_2O , PdO , Rh_2O_3 , and Ru_2O_3), and the semi-soluble components in the waste (P_2O_5 and SO_4).

7.7 BALANCE OF PROCESS

Some components are likely to have a significant effect on the operability of the HLW immobilization facility. Offgas and emissions control systems could be affected by the amount of mercury, cadmium, and other volatile components in the waste while chlorides and sulfates could contribute to the general corrosion of offgas system components. Criticality controls will undoubtedly be effected by the actinides in the waste.

Approximately 40 p/M mercury may exist in Tank 241-AY-102 heel and 42 to 327 p/M mercury in Tank 241-C-106 waste. On a dried solids basis, as much as 0.017 wt% mercury (173 p/M) could be present in the combined Phase I feed from Tanks 241-C-106 and 241-AY-102. Mercury is relatively insoluble in borosilicate glass and tends to volatilize in the melter depending on cold cap coverage, plenum temperature, and feed/glass composition. Melter studies have indicated that as much as 50 percent of the mercury in the melter feed could volatilize in the melter. Based on the observed product splits at the SRS plant, it is anticipated that 85 percent of the mercury could be removed during the feed preparation process (with formic acid treatment) and 15 percent distributed to the melter. If this distribution also reflects the behavior of mercury in the Phase I immobilization facility, the melter feed concentration could vary from 6.3 to 49 p/M mercury (for Tanks 241-C-106 and 241-AY-102 waste). On this basis, mercury emissions from the melter (rated at 1.0 MT/day of glass @ 35 percent WOL) could range from 0.0022 to 0.0172 kg/day with an average value of 0.0095 kg/day. At SRS, the anticipated emission rate is estimated to be 0.136 kg/day of mercury from the melter offgas condenser. The HLW flowsheet currently assumes that all the mercury removed from the feed will be purified and shipped to SRS for disposal.

Less than 3 percent of the cadmium that volatilizes in the melter is likely to precipitate or be incorporated into the deposits in the melter offgas line. The solubility limit of CdO in borosilicate glass is estimated to be 2.5 wt%, while the CdO DF across the melter is about 350. Thus, most of the CdO is likely to be incorporated into the glass product. Based on these estimates and the projected CdO composition of Tanks 241-AZ-101 and 241-AZ-102 waste (1.80 wt% and 3.54 wt%, respectively), offgas line deposits are unlikely to occur if Tanks 241-AZ-101 and 241-AZ-102 wastes are blended (2.84 wt% CdO). On the other hand, long term problems could arise if Tank 241-AZ-102 waste is processed as a separate feed and mechanical systems are not available to remove potential deposits from the melter offgas line. While CdO deposits may be a contributing factor, sodium metaborate and chloride salt deposits are likely to be a much more significant problem with respect to the melter offgas line.

Most of the chloride, fluoride, and sulfate should be separated from the HLW during the sludge-washing process. However, salt deposits that consist of alkali-rich chlorides, sulfates, borates, and fluorides are likely to form in the melter offgas line. The design specifications for the HWVP were based on an assumed 40-year operating life. The applicable corrosion limits were set at less than 125 μ /year. Depending on pH, several materials were identified as candidate materials for use in a high-halide environment. Hastelloy C-22 exhibited the lowest corrosion rates, generally less than 125 μ /year. The bounding case for chloride appears to be a blend of tanks 241-C-106 and 241-AY-102 with an estimated chloride concentration of 705 p/M in the melter and 1,400 p/M in the melter feed tank. These concentrations are bounded by the simulant offgas condensate chloride concentrations used in the materials tests for HWVP. The same general conclusion also applies to Tank 241-AY-102 wastes, which is the limiting case for fluoride. For sulfate, the limiting case is Tank 241-AZ-101 waste, where the maximum sulfate concentration is 115 p/M in the melter (@ 38 percent WOL). This may be compared to a limiting concentration of 2,500 p/M sulfate in the WVDP melter, that was found to be completely acceptable with respect to electrode corrosion.

Nitrogen oxide emissions could be on the order of 12 to 18 MT/year when processing NCAW feed through the vitrification process. If nitrogen oxide emissions exceed 40 MT/year, air quality regulations could require use of BACT, that would be determined on a case-by-case basis. Selective catalytic reduction of nitrogen oxides with ammonia may be the preferred process for BACT control. Nitric oxide emissions and NO_x treatment processes are identified in the TWRS flowsheet (Orme 1995).

The plutonium inventory in the Phase I waste (Tanks 241-AZ-101/241-AZ-102/241-C-106/241-AY-102) is estimated to be at most 173 kg PuO₂ with 90 percent as ²³⁹Pu and 8 percent as ²⁴⁰Pu. The most limiting case is Tank 241-AY-102 waste based on updated core sample information where the maximum PuO₂ concentration is 0.03 wt% in the melter feed (Table 6-27). At SRS, the solubility of PuO₂ was determined to be 4 wt% in Frit 21 and 7.0 wt% in Frit 411 (Plodinec 1979, Soper 1982). Both frits are essentially borosilicate glasses with different concentrations of Na₂O, Li₂O, and TiO₂. Because the solubility of PuO₂ in borosilicate glass appears to be at least 140 times the limiting concentration in NCAW waste, the precipitation of PuO₂ as a separate phase is not anticipated. The possibility of a criticality incident in the DWPF melter due to the accumulation of PuO₂ or plutonium metal has been examined at SRS (Soper 1982). The solubility of plutonium was found to greatly exceed the amount present in SRS's waste. A thermodynamic analysis of PuO₂ reduction shows that it will be one of the last oxides in the glass to reduce to metal. In particular, iron, nickel, and manganese oxides will be reduced to metal before PuO₂ is reduced. Large amounts of iron and other metals in the melter would cause severe operating problems and shorting of the electrodes before there would be any significant accumulation of plutonium.

Criticality control in the melter should be based on the possible accumulation of a plutonium-rich secondary phase at the bottom of the melter. Because the British have observed that cerium tends to form insoluble precipitates with actinides (plutonium and neptunium) in molten glass, the possibility of cerium, uranium, or some other carrier-induced precipitation of plutonium needs to be examined. If glass formulation testing shows that plutonium-rich crystals will not form or precipitate in the melter, criticality control should be based on dilution of plutonium in the glass such that K-effective would be much less than 0.95.

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