

ADVANCED HIGH-TEMPERATURE, HIGH-PRESSURE TRANSPORT REACTOR GASIFICATION

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ADVANCED HIGH-TEMPERATURE, HIGH-PRESSURE TRANSPORT REACTOR GASIFICATION

ABSTRACT

The transport reactor development unit (TRDU) was modified to accommodate oxygen-blown operation in support of a Vision 21-type energy plex that could produce power, chemicals, and fuel. These modifications consisted of changing the loop seal design from a J-leg to an L-valve configuration, thereby increasing the mixing zone length and residence time. In addition, the standpipe, dipleg, and L-valve diameters were increased to reduce slugging caused by bubble formation in the lightly fluidized sections of the solid return legs. A seal pot was added to the bottom of the dipleg so that the level of solids in the standpipe could be operated independently of the dipleg return leg. A separate coal feed nozzle was added that could inject the coal upward into the outlet of the mixing zone, thereby precluding any chance of the fresh coal feed back-mixing into the oxidizing zone of the mixing zone; however, difficulties with this coal feed configuration led to a switch back to the original downward configuration. Instrumentation to measure and control the flow of oxygen and steam to the burner and mix zone ports was added to allow the TRDU to be operated under full oxygen-blown conditions.

In total, ten test campaigns have been conducted under enriched-air or full oxygen-blown conditions. During these tests, 1515 hours of coal feed with 660 hours of air-blown gasification and 720 hours of enriched-air or oxygen-blown coal gasification were completed under this particular contract. During these tests, approximately 366 hours of operation with Wyodak, 123 hours with Navajo subbituminous coal, 143 hours with Illinois No. 6, 106 hours with SUFCo, 110 hours with Prater Creek, 48 hours with Calumet, and 134 hours with a Pittsburgh No. 8 bituminous coal were completed. In addition, 331 hours of operation on low-rank coals such as North Dakota lignite, Australian brown coal, and a 90:10 wt% mixture of lignite and wood waste were completed. Also included in these test campaigns was 50 hours of gasification on a petroleum coke from the Hunt Oil Refinery and an additional 73 hours of operation on a high-ash coal from India. Data from these tests indicate that while acceptable fuel gas heating value was achieved with these fuels, the transport gasifier performs better on the lower-rank feedstocks because of their higher char reactivity.

Comparable carbon conversions have been achieved at similar oxygen/coal ratios for both air-blown and oxygen-blown operation for each fuel; however, carbon conversion was lower for the less reactive feedstocks. While separation of fines from the feed coals is not needed with this technology, some testing has suggested that feedstocks with higher levels of fines have resulted in reduced carbon conversion, presumably due to the inability of the finer carbon particles to be captured by the cyclones. These data show that these low-rank feedstocks provided similar fuel gas heating values; however, even among the high-reactivity low-rank coals, the carbon conversion did appear to be lower for the fuels (brown coal in particular) that contained a significant amount of fines. The fuel gas under oxygen-blown operation has been higher in hydrogen and carbon dioxide concentration since the higher steam injection rate promotes the water-gas shift reaction to produce more CO₂ and H₂ at the expense of the CO and water vapor. However, the high water and CO₂

partial pressures have also significantly reduced the reaction of hydrogen sulfide with the calcium-based sorbents and thus the capture of sulfur in the circulating-bed material.

Since warm-gas cleanup is utilized, the unconverted steam and coal moisture injected into the gasifier will remain in the fuel gas entering the gas turbine. When the air-blown and oxygen-blown fuel gas heating values are compared for the wet product gas streams, it is apparent that only a slight improvement in product gas heating value entering the gas turbine is achieved with oxygen-blown operation. In order to keep the gas turbine firing temperature down to prevent thermal NO_x formation, typically large amounts of nitrogen or steam are injected into the gas turbine combustor such that the fuel gas heating is typically not much greater than 115 Btu/scf as-fired. In essence, the transport reactor has either injected the nitrogen with the oxidant (in the form of air) into the gasifier instead of directly into the gas turbine combustor in air-blown mode or has injected the steam directly into the gasifier instead of the gas turbine combustor in the oxygen-blown case. However, in a Vision 21 plant, where chemicals or fuel production are being considered and where potentially conventional cold-gas cleanup technology would be utilized to remove the water vapor from the fuel gas stream, significantly higher concentrations of desirable fuel gas constituents are achieved with oxygen-blown operation.

The TRDU and hot-gas filters have operated for over 2175 hours in gasification mode and over 2500 hours total with no major candle failures. The candles have exhibited no significant loss in candle permeability. The baseline “cleaned” filter differential pressure typically increased from 20 to approximately 80 inches H_2O over the course of most tests. The inlet particulate loading has ranged from approximately 3500 to 33,800 ppm, with the filter ash averaging between 20 to 70 wt% carbon with a low bulk density around 20 lb/ft^3 . The average filter ash particle size has ranged from approximately 7 to 22 μm in size and was essentially representative of the coal ash from very early in the gasification test. The initial rapid recovery of the filter differential pressure along with the small size, the lack of cohesiveness seen in other filter ashes, and the low density of the ash had suggested that a high percentage of the filter cake would be reentrained back onto the filters after they are backpulsed. The large increase in filter baseline differential pressure also suggests that a thin but low-porosity (permeable) filter cake is remaining on the surface of the candle and is not being removed during backpulsing. The low bulk density and high flowability of the filter ash possibly suggests that the inlet ash is able to move or shift on the surface of the candle to reach some optimum (minimum) porosity, leading to low gas permeability across the candle.

Continuous measurement of mercury in the warm fuel gas has been another goal of the project. After considerable trial and error, a fuel gas-conditioning system and Hg continuous emission monitor (CEM) analyzer has been configured to allow the continuous measurement of mercury emissions. Sampling issues for both the wet-chemistry and Hg CEM techniques have been resolved, so that good agreement between the two techniques is being achieved. Wet-chemistry analysis has shown the mercury to essentially be in the elemental form. The EERC continues to utilize advanced scanning electron microscopy (SEM) techniques where appropriate to determine the chemistry of any bed material agglomeration or deposition samples. No high levels of reactive sulfide have been measured in any TRDU samples that would make the residual solids a hazardous waste.

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ADVANCED HIGH-TEMPERATURE, HIGH-PRESSURE TRANSPORT REACTOR GASIFICATION

1.0 OBJECTIVES

The objectives of the advanced high-temperature, high-pressure transport reactor gasification system with the transport reactor demonstration unit (TRDU) located at the Energy & Environmental Research Center (EERC) is to demonstrate and optimize the performance of the transport reactor gasification concept in a pilot-scale system prior to longer-term demonstration tests at the Power Systems Development Facility (PSDF). The primary focus of the experimental effort over the last 6 years has been to modify the TRDU and conduct oxygen-blown gasification testing including investigating the effects of coal type on gasification performance. A secondary objective of the project has been the testing of hot-gas filter element performance (particulate collection efficiency, filter pressure differential, filter cleanability, and durability) as a function of temperature and filter face velocity during relatively short-term operation (100–200 hours). The filter vessel is used in combination with the TRDU to evaluate the performance of selected hot-gas filter elements under gasification operating conditions. This work directly supports the PSDF utilizing the Kellogg Brown & Root (KBR) transport reactor located at Wilsonville, Alabama (1).

2.0 BACKGROUND INFORMATION

The U.S. Department of Energy (DOE) National Energy Technology Laboratory (NETL) has a gasification program that has made gasification one of the centerpiece technologies being developed for future power, fuels, and chemicals production under the Vision 21 Program. In order to economically make fuels and chemicals from synthesis gas, the fuel should have a minimal amount of nitrogen in the fuel gas in order to minimize the size of the downstream unit operations required to produce the desired fuel or chemical slate.

The Gas Cleanup Program is intended to develop and demonstrate gas stream cleanup options for use in combustion- or gasification-based advanced power systems. One objective of the NETL gas cleanup program is to support the development and demonstration of barrier filters to control particulate matter. The goal is not only to meet current New Source Performance Standards (NSPS) with respect to particulate emissions, but also to protect high-efficiency gas turbines and control particulate emissions to low enough levels to meet more stringent regulatory requirements anticipated in the future. DOE NETL is investing significant resources in the PSDF under a Cooperative Agreement with Southern Company Services, Inc. (SCS). The Wilsonville facility will include three modules, including an advanced gasifier module, a gas cleanup module, and a combustor/turbine module. The gasifier module incorporates the KBR transport reactor technology for both gasification and combustion (1).

The TRDU was built and operated at the EERC under Contract No. C-92-000276 with SCS. KBR designed and procured the reactor and provided valuable on-site personnel for start-up and during operation. The Electric Power Research Institute (EPRI) was involved in establishing the program and operating objectives with the EERC project team.

The purpose of the previous program was to build a reactor system larger than the TRTU located in Houston, Texas, in support of the Wilsonville PSDF transport reactor train. The program was to address design and operation issues for the Wilsonville unit and also help develop information on the operation of the unit to decrease start-up costs.

The TRDU (design rate 240-lb/hr coal–limestone feed rate) provides an intermediate scale to the TRTU (up to 10-lb/hr coal–limestone feed rate) and the Wilsonville transport reactor (3400-lb/hr feed rate). Some of the design, construction, start-up, and operational issues for the Wilsonville transport train are being addressed during this project.

The four major design criteria that were established by EPRI were met (2): coal feed rate, operating pressure, carbon conversion, and high heating value of the product gas. Major accomplishments included showing that the TRDU performed well hydrodynamically, that it had the ability to switch from combustion mode to gasification mode easily and safely, that solids could be fed to and removed from the system, and that the J-leg/standpipe and cyclone performed according to their design specifications.

3.0 PROJECT DESCRIPTION

3.1 TRDU

The TRDU has an exit gas temperature of up to 980°C (1800°F), a gas flow rate of 325 scfm, and an operating pressure of 120–150 psig. The TRDU system can be divided into three sections: the coal feed section, the TRDU, and the product recovery section. The TRDU proper, as shown in Figure 1, consists of a riser reactor with an expanded mixing zone at the bottom, a disengager (which is an actual cyclone, unlike the disengager at the PSDF), and a primary cyclone and standpipe and dipleg under cyclone for recycling the bed material back to the mixing zone. The standpipe is connected to the mixing section of the riser by a J-leg transfer line. All of the components in the system are refractory-lined and designed mechanically for 150 psig and an internal temperature of 1090°C (2000°F). Table 1 summarizes the operational performance for the TRDU under the previous test program (3).

The premixed coal and limestone feed to the transport reactor can be admitted through three nozzles, which are at varying elevations. Two of these nozzles are located near the top of the mixing zone (gasification), and the remaining one is near the bottom of the mixing zone (combustion). During operation of the TRDU, feed is admitted through only one nozzle at a time. The coal feed is measured by an rpm-controlled metering auger. Oxidant is fed to the reactor through two pairs of nozzles at varying elevations within the mixing zone. For the combustion mode of operation, additional nozzles are provided in the riser for feeding secondary air. Hot solids from the standpipe are circulated into the mixing zone, where they come into contact with the nitrogen and the steam being injected into the J-leg. This feature enables spent char to contact steam prior to the fresh

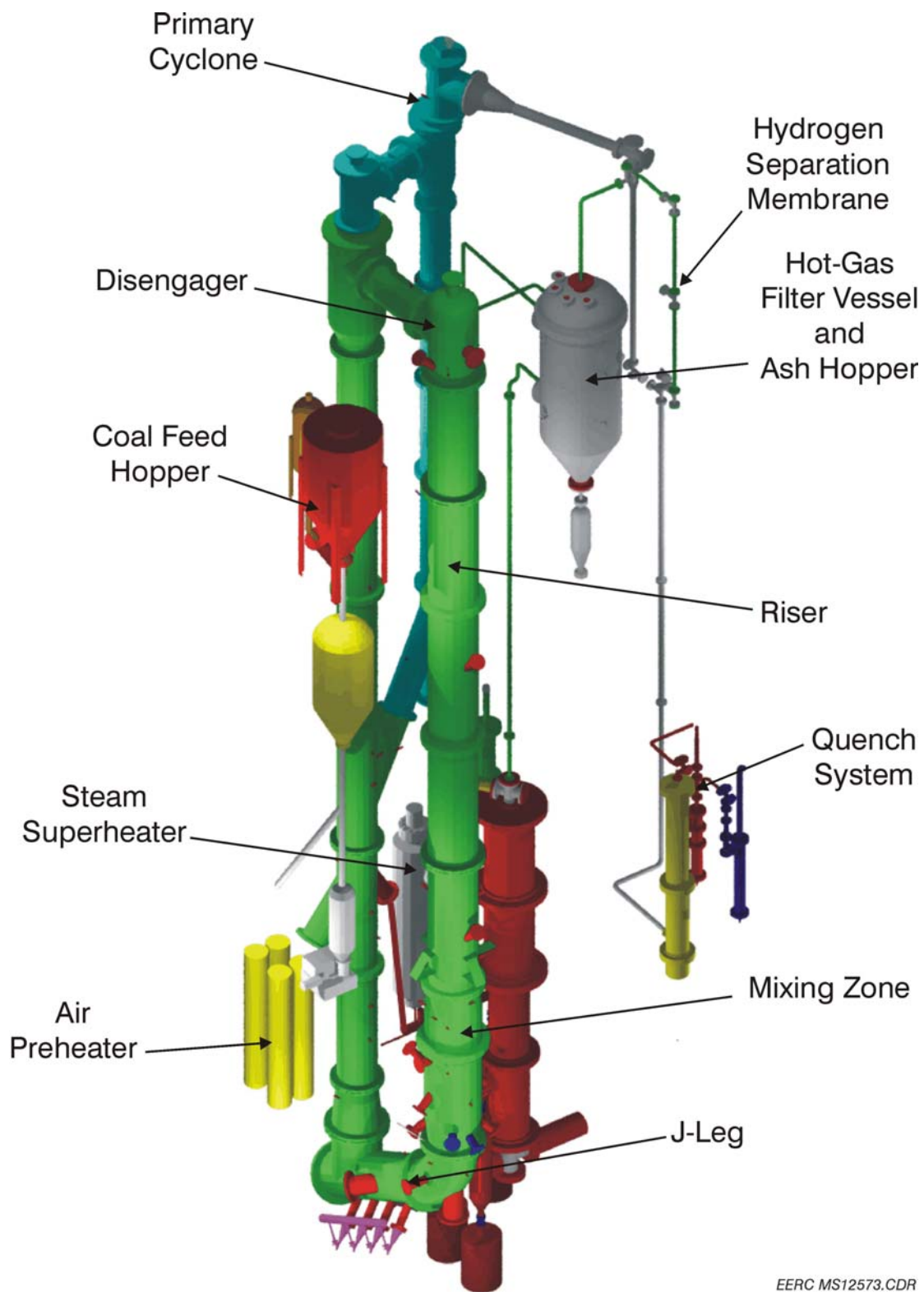


Figure 1. TRDU and hot-gas vessel in the EERC gasification tower.

Table 1. TRDU Design and Typical Actual Operating Conditions

Parameter	Design	P056 and P057	P056	P057
Conditions	Gasification	Gasification	Gasification	Gasification
Coal	Illinois No. 6	Wyodak	Illinois No. 6	SUFCo
Moisture Content, %	5	20	8.5	9.5
Pressure, bar	9.3	9.3	9.3	9.3
Steam:Coal Ratio, lb/lb coal	0.34	0.29	0.39	0.14 to 0.41
Air:Coal Ratio, lb/lb coal	4.0	2.69	2.59	3.34–3.45
Ca:S Mole Ratio, sorbent	1.5	2	2	2
Coal Feed Rate, lb/hr	198	276.6	232.5	220
J-Leg Zone, °C, avg.	9.931010e+13	8.00850841e+17	9.019359e+17	866–876
Mixing Zone, °C, avg.				920–950
Riser, °C, avg.				894–914
Standpipe, °C, avg.				828–860
Dipleg, °C, avg.				555–591
TRDU Outlet, °C, avg.				856–877
Carbon Conversion, %	>80	89	76	72–87
Carbon in Bed, %, Standpipe	20 to 40	6 to 15	6 to 15	5 to 20
Riser Velocity, ft/s	31.3	30	24	25–31
Standpipe Velocity, ft/s	0.1	0.4 to 0.5	0.45	0.4–0.45
Circulation Rate, lb/hr	30000	3000 to 6000	4000	2650–4200
HHV of Fuel Gas, Act., Btu/scf,	100	62–75	61113	52–75
Cor., Btu/scf		105–117		93–130
Duration, hr	NA	179	41	118

coal feed. This staged gasification process is expected to enhance the process efficiency. Gasification or combustion and desulfurization reactions are carried out in the riser as coal, sorbent, and oxidant (with steam for gasification) flow up the reactor. The solids circulation into the mixing zone is controlled by the solids level in the standpipe and by the gas flow rates and distribution in the J-leg aeration nozzles.

The riser, disengager, standpipe, and cyclones are equipped with several internal and skin thermocouples. Nitrogen-purged pressure taps are also provided to record differential pressure across the riser, disengager, and the cyclones. The data acquisition and control system scans the data points every one-half second and saves the process data every 30 seconds. The bulk of entrained solids leaving the riser is separated from the gas stream in the disengager and circulated back to the riser via the standpipe. A solids stream can be withdrawn from the standpipe via an auger to maintain the system's solids inventory. Gas exiting the disengager enters a primary cyclone. Solids from the primary cyclone were collected in a lock hopper for earlier tests through approximately Test P055. In tests after P055, the dipleg solids have been recirculated back to the standpipe through the dipleg crossover. Gas exiting this cyclone enters a jacketed-pipe heat exchanger before entering the hot-gas filter vessel (HGFV). The cleaned gases leaving the HGFV enter a quench system before being depressurized and vented to a flare.

The quench system uses a sieve tower and two direct-contact water scrubbers to act as heat sinks and remove impurities. All water and organic vapors are condensed in the first scrubber, with

the second scrubber capturing entrained material and serving as a backup. The condensed liquid is separated from the gas stream in a cyclone that also serves as a reservoir. Liquid is pumped either to a shell-and-tube heat exchanger for reinfection into the scrubber or down to the product receiver barrels.

3.2 Hot-Gas Filter Vessel

This vessel is designed to handle all of the gas flow from the TRDU at its expected operating conditions. The vessel is approximately 48-in. ID and 185 in. long and is designed to handle gas flows of approximately 325 scfm at temperatures up to 980°C (1800°F) and 130 psig. The refractory has a 28-in. ID with a shroud diameter of approximately 22 in. The vessel is sized such that it could handle candle filters up to 1.5 m long; however, 1-m candles were utilized in the 1000°F (540°C) gasification tests. Candle filters are 2.375-in. OD with 4-in. center line-to-center line spacing. The filter design criteria are summarized in Table 2, and a schematic is given in Figure 2.

The total number of candles that can be mounted in the current geometry of the tube sheet is 19. This enables filter face velocities as low as 2.5 ft/min to be tested using 1-m candles. Tests consisted of 200-hr hot-gas filter tests under gasification conditions using the TRDU with the filter operating at temperatures of 540°–650°C (1000°–1200°F) and 120 psig. Higher face velocities would be achieved by using fewer candles. The test program performed the first filter test at 540°–650°C (1000°–1200°F), 120 psig, and 2.75 ft/min face velocity. All subsequent testing was performed after removing six candles to increase the face velocity to approximately 4.0 to 4.5 ft/min at the same operating temperature and pressure. The openings for the six removed candles were blanked off. This program has tested an Industrial Filter & Pump (IF&P) ceramic tube sheet and

Table 2. Design Criteria and Typical Actual Operating Conditions for the Pilot-Scale Hot-Gas Filter Vessel

Operating Conditions	Design	Actual
Inlet Gas Temperature	540°–980°C	520°–580°C
Operating Pressure	150 psig	120 psig
Volumetric Gas Flow	325 scfm	350 scfm
Number of Candles	19 (1 or 1.5 meter)	13 (1-meter)
Candle Spacing	4 in. C to C	4 in. C to C
Filter Face Velocity	2.5–10 ft/min	4.5 ft/min
Particulate Loading	<10,000 p.m.	< 7,000 p.m.
Temperature Drop Across HGFV	<30° C	25°C
Nitrogen Backpulse System Pressure	up to 800 psig	250 to 350 psig
Backpulse Valve Open Duration	up to 1-sec duration	½-sec duration

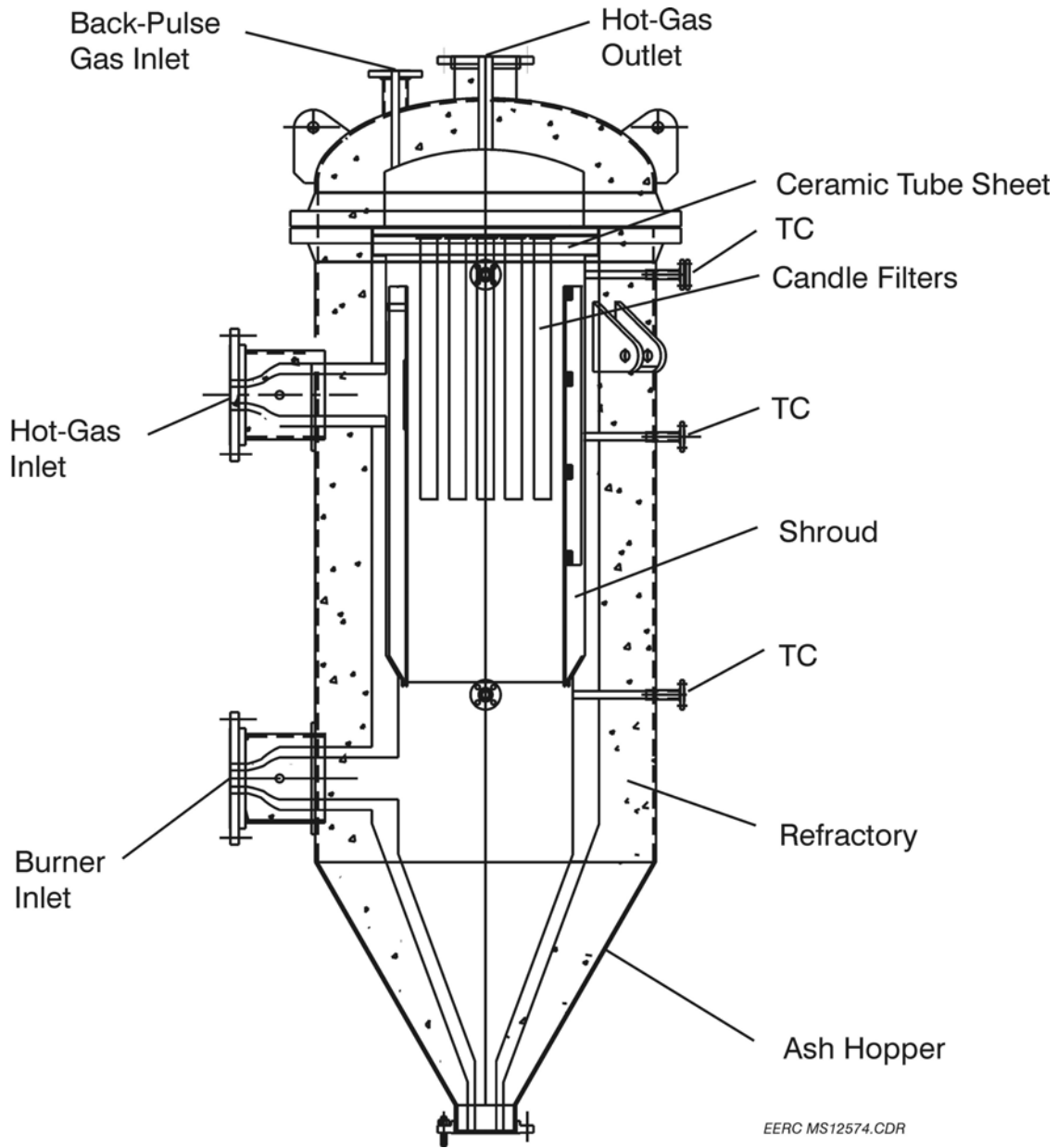


Figure 2. Schematic of the filter vessel design with internal refractory, tube sheet, and shroud.

Fibrosic candles, silicon carbon-coated ceramic fiber candles from the 3M Company, along with sintered metal (iron aluminide) and Vitropore silicon carbon ceramic candles from Pall Advanced Separation Systems Corporation. Later tests also utilized a metal tube sheet manufactured with expansion cones to allow for thermal stresses. Since the metal tube sheet was installed, candle filter fail-safes from Westinghouse Science and Technology Center have also been tested.

The ash letdown system consists of two sets of alternating high-temperature valves with a conical pressure vessel to act as a lock hopper. Additionally, a preheat natural gas burner attached

to a lower inlet nozzle on the filter vessel can be used to preheat the filter vessel separately from the TRDU. The hot gas from the burner enters the vessel via a nozzle inlet separate from the dirty gas.

The high-pressure nitrogen backpulse system is capable of backpulsing up to four sets of four or five candle filters with ambient-temperature nitrogen in a time-controlled sequence. The pulse length and volume of nitrogen displaced into the filter vessel is controlled by regulating the pressure (up to 800 psig) of the nitrogen reservoir and the solenoid valves used to control the timing of the gas pulse. Figure 1 also shows the filter vessel location and process piping in the EERC gasifier tower. Since all the filter tests are to be completed in the 540°–650°C (1000°–1200°F) range, a length of heat exchanger was used to drop the gas temperature to the desired range. In addition, sample ports both upstream and downstream of the filter vessel have been utilized for obtaining particulate and hazardous air pollutant (HAP) samples.

4.0 ACCOMPLISHMENTS

4.1 TRDU Fuel Analysis

The fuels tested in the TRDU have been a Powder River Basin (PRB) subbituminous coal from the Wyodak seam at the Belle Ayr Mine in Gillette, Wyoming; an Illinois No. 6 bituminous coal from Seam 6 of the Creek Palm Mine near Mirressa, Illinois; a western bituminous coal mined from the Hiawatha seam at the SUFCo Mine in Salina, Utah; a bituminous coal from the Prater Creek Mine in eastern Kentucky; a bituminous coal from Mary Lee seam at the Calumet Mine in Alabama; a bituminous coal from Pittsburgh No. 8 seam from Consul's Bailey Mine; a petroleum coke from the Hunt Oil Refinery in Tuscaloosa, Alabama; a high-ash subbituminous coal from the Navajo Mine in the Four Corners region of New Mexico; and three different North Dakota lignites. Wood residue hog fuel was coal-fed with one of the lignites. Tables 3 and 4 shows the proximate, ultimate, and x-ray fluorescence (XRF) analysis of these fuels. In addition, through an intergovernment agreement between the Australian government and the U.S. government and separately through U.S. AID, three foreign coals (two different Australian brown coals and an as-received and washed high ash coal from India) have also been tested in the TRDU. Table 5 shows the proximate and ultimate analysis for these fuels also. Table 6 shows the XRF and loss on ignition (LOI) analyses for the Plum Run dolomite and the Montana and Longview limestones utilized in these tests. All fuels were mixed with calcium-based sorbents to provide a Ca/S molar ratio of approximately 1.5 to 2 on a sorbent-only basis for the fuels being gasified. Figure 3 shows the particle-size distribution for the coals tested on the TRDU. In general, the coal mean feed size is between approximately 300 to 500 μm which is larger than the circulating bed material mean size of 200 μm . Because of the friability of the brown coals, significantly more fines were present in the feed coal than other coals.

Table 3. Proximate, Ultimate, HHV, and XRF Analysis Results for TRDU Testing

	-10-mesh Wyodak Subbituminous Coal	-10-mesh Illinois No. 6 Bituminous Coal	-10-mesh SUFCo Bituminous Coal	-10-mesh Center Lignite Coal	-10-mesh Falkirk Lignite Coal	-10-mesh Freedom Lignite Coal	-1/8" Wood Hog Fuel
Proximate Analysis, as run, wt%							
Moisture	20.0	8.5	9.5	35.5	29.50	26.80	12.2
Volatile Matter	38.9	36.0	39.1	24.3	30.92	32.52	73.1
Fixed Carbon	36.4	44.8	43.8	25.3	27.89	32.48	11.1
Ash	4.7	10.7	7.6	14.87	11.69	8.2	3.5
Ultimate Analysis, MF, ¹ wt%							
Carbon	69.06	69.27	77.10	56.72	58.64	62.61	48.36
Hydrogen	5.19	5.03	4.61	4.05	4.04	4.25	5.76
Nitrogen	0.84	1.1	1.29	0.80	0.81	0.96	0.62
Sulfur	0.44	3.55	0.36	1.2	1.06	0.94	0.16
Oxygen	18.63	9.34	8.29	19.68	18.87	20.05	41.07
Ash	5.85	11.7	8.4	23.1	16.58	11.20	4.0
Ash Composition, % as oxides							
Calcium, CaO	26.6	3.2	16.3	8.3	15.5	15.9	51.6
Magnesium, MgO	7.0	1.6	3.0	2.8	8.9	5.5	5.4
Sodium, Na ₂ O	1.3	1.1	4.6	1.8	0.7	6.0	3.5
Silica, SiO ₂	27.8	53.9	38.3	48.3	41.3	34.6	22.7
Aluminum, Al ₂ O ₃	13.1	21.2	9.3	14.2	12.8	12.6	2.7
Ferric, Fe ₂ O ₃	5.5	13.6	6.1	6.8	4.5	6.6	2.2
Titanium, TiO ₂	1.3	0.9	0.8	0.6	0.5	0.3	0.2
Phosphorus, P ₂ O ₅	1.0	0.2	0.2	0.0	0.2	0.5	2.9
Potassium, K ₂ O	0.3	1.9	0.2	2.0	0.4	0.3	7.9
Sulfur, SO ₃	16.0	2.5	21.1	2.2	14.3	17.6	0.9
High Heating Value							
Moisture-Free, Btu/lb	11,700	12,080	12,200	9446	9963	10,669	8,089
As-Received, Btu/lb	9750	11,300	11,040	6093	7024	7810	7,102

¹ Moisture-free.

Table 4. Proximate, Ultimate, HHV, and XRF Analysis Results for TRDU Testing

	-10-mesh Tuscaloosa Petroleum Coke	-10-mesh Prater Creek Bituminous Coal	-10-mesh Navajo Subbituminous	-10-mesh Pittsburgh No. 8 Bituminous Coal	-10-mesh Calumet Bituminous Coal
Proximate Analysis, as run, wt%					
Moisture	0.9	7.0	10.0	2.2	3.3
Volatile Matter	9.6	38.9	33.5	37.8	32.3
Fixed Carbon	88.5	47.5	35.5	52.6	49.1
Ash	1.0	6.6	21.0	7.4	15.7
Ultimate Analysis, MF, wt%					
Carbon	90.7	76.2	58.5	77.9	66.7
Hydrogen	3.9	4.8	4.5	5.3	4.3
Nitrogen	1.7	1.6	1.2	1.4	1.9
Sulfur	5.5	0.8	1.1	1.6	0.7
Oxygen	0.0	9.4	11.3	6.3	10.3
Ash	1.0	7.1	23.3	7.5	16.1
Ash Composition, % as oxides					
Calcium, CaO	11.9	1.8	3.5	3.3	0.8
Magnesium, MgO	5.1	1.2	1.4	0.9	2.2
Sodium, Na ₂ O	1.0	0.4	1.4	0.2	0.1
Silica, SiO ₂	18.9	54.4	58.4	52.4	58.5
Aluminum, Al ₂ O ₃	4.8	30.1	25.5	24.3	28.2
Ferric, Fe ₂ O ₃	7.6	6.9	6.2	13.7	5.1
Titanium, TiO ₂	0.0	1.0	1.2	1.0	1.4
Phosphorus, P ₂ O ₅	0.1	0.3	0.1	0.5	0.3
Potassium, K ₂ O	0.7	0.8	0.5	1.6	2.5
Sulfur, SO ₃	13.8	2.0	2.6	1.9	0.8
Vanadium, V ₂ O ₅	30.2	ND	ND	ND	ND
Nickel, NiO	6.0	ND	ND	ND	ND
High Heating Value					
Moisture-Free, Btu/lb	12,080	13,813	9777	13,627	12,214
As-Received, Btu/lb	11,300	12,847	8880	13,327	11,809

Table 5. Proximate, Ultimate, HHV, and XRF Analyses of Australian Brown Coals and High-Ash Indian Subbituminous A Coals Utilized in Tests P075 and P077

	-10-mesh Dried Loy Yang Brown Coal	-10-mesh Dried Lochiel Brown Coal	-10-mesh Raw Indian Subbituminous A Coal	-10-mesh Washed Indian Subbituminous A Coal
Proximate Analysis, as run,				
wt%	15.0	18.0	4.7	9.0
Moisture	48.7	43.8	26.2	26.1
Volatile Matter	35.3	25.9	30.5	29.3
Fixed Carbon	0.9	12.3	38.6	35.5
Ash				
Ultimate Analysis, MF, wt%				
Carbon	65.4	56.1	46.2	44.9
Hydrogen	4.6	4.3	3.32	3.16
Nitrogen	0.8	0.7	1.1	1.1
Sulfur	0.4	3.6	0.55	0.61
Oxygen	27.7	20.4	8.32	11.2
Ash	1.1	15.0	40.5	39.1
Ash Composition, % as				
oxides	6.9	11.8	1.2	1.5
Calcium, CaO	13.2	10.4	0.6	0.6
Magnesium, MgO	10.3	9.1	0.4	0.5
Sodium, Na ₂ O	26.0	27.9	61.5	60.4
Silica, SiO ₂	8.4	6.5	27.9	28.5
Aluminum, Al ₂ O ₃	10.4	4.7	4.5	5.1
Ferric, Fe ₂ O ₃	0.7	0.8	2.4	2.1
Titanium, TiO ₂	0.1	0.0	0.6	0.1
Phosphorus, P ₂ O ₅	1.4	0.4	1.0	1.1
Potassium, K ₂ O	22.7	28.4	0.0	0.0
Sulfur, SO ₃	ND	ND	0.170	0.162
Mercury, µg/g	ND	ND		
Higher Heating Value				
MF, Btu/lb	11,112	9011	6864	7218
As-Received, Btu/lb	9445	7389	6555	6568

Table 7 shows the ASTM coal classification scheme. Except for the petroleum coke, coals that have been tested in the TRDU have ranged as high as a high-volatile B bituminous coals or lower. The transport reactor technology has been considered to be more suited to lower-rank coals that have a higher char reactivity; however, a number of higher-rank bituminous coals were also tested to determine their performance in a transport reactor.

Table 6. XRF Analyses of Plum Run Dolomite and Longview and Montana Limestones

	-35-mesh Plum Run Dolomite (PRD)	-35-mesh Longview Limestone (LVLS)	-35-mesh Montana Limestone (MLS)
Sorbent Composition, % as oxides			
Calcium, CaO	66.6	90.1	73.6
Magnesium, MgO	27.5	5.6	0.4
Sodium, Na ₂ O	0.3	0.0	0.0
Silicon, SiO ₂	2.7	2.0	25.3
Aluminum, Al ₂ O ₃	1.0	0.2	0.0
Ferric, Fe ₂ O ₃	1.3	0.2	0.0
Titanium, TiO ₂	0.0	0.0	0.0
Phosphorus, P ₂ O ₅	0.0	0.0	0.0
Potassium, K ₂ O	0.3	0.3	0.3
Sulfur, SO ₃	0.4	0.4	0.4
Loss on Ignition, as run	43.1	ND	36.6

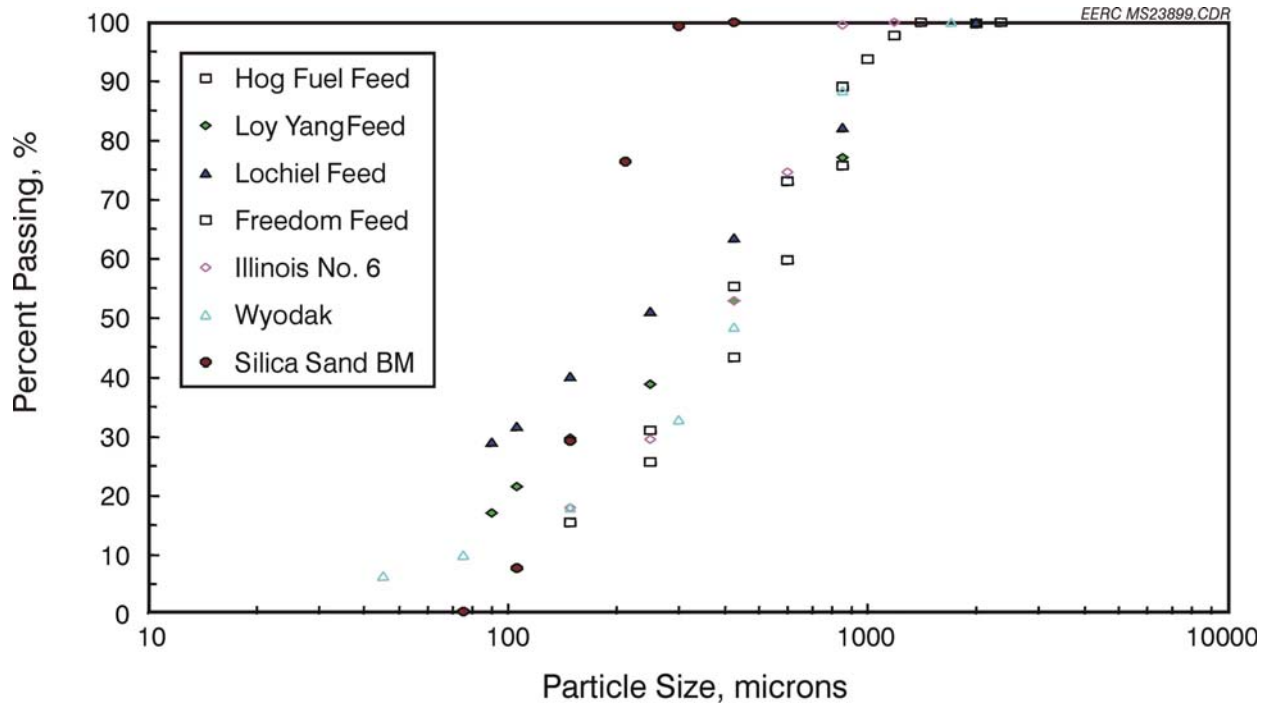


Figure 3. Particle-size distribution of feed coals tested in the TRDU.

Table 7. ASTM International Coal Classification Criteria

Class	Group	Fixed Carbon ^a	Volatile Matter ^a	Heating Value
Anthracitic	Metaanthracite	>98	<2	
	Anthracite	92–98	2–8	
	Semianthracite	86–92	8–14	
Bituminous	Low-volatile	78–86	14–22	
	Medium-volatile	69–86	22–31	
	High-volatile A	<69	>31	>14,000
	High-volatile B			13,000–14,000
	High-volatile C			10,500–13,000
Subbituminous	Subbituminous A			10,500–11,500
	Subbituminous B			9500–10,5000
	Subbituminous C			8300–9500
Lignitic	Lignite A			6300–8300
	Lignite B			<6300

Note: This classification system is based on ASTM Standard D 388–66, which is published annually by ASTM in its compilation of standards.

^a The fixed carbon and volatile matter, reported as percentages, are determined on a dry, mineral-matter-free basis. The mineral matter is calculated from the ash content by the Parr formula: mineral matter = 1.08(percent ash + 0.55 [percent sulfur]).

^b Calculated on mineral-matter-free coal with bed moisture content.

4.2 TRDU Testing with the J-Leg Loop Seal

As modifications to the TRDU were being contemplated and then designed, three additional air-blown and oxygen-enriched tests were completed utilizing the original J-leg configuration.

A TRDU test campaign was conducted during the weeks of March 1–11, 1999, that generated 138 hours of coal feed and 107 hours of operation in coal gasification mode with the system gases and fly ash passing through the filter vessel during the whole test campaign. These tests were terminated early because of deposition problems in the mixing zone with the SUFCo fuel and solids flow problems from both the disengager and primary cyclone cones back into the standpipe or dipleg with the petroleum coke test.

4.2.1 TRDU Gasification Tests P060 and P061

TRDU gasification Test P060 was an air-blown test conducted over the period of March 1 – 5, 1999, utilizing SUFCo coal. This test was to compare the gasifier performance after the TRDU had been modified by enlarging the diameter of the mixing zone to increase the solids residence time and decrease the gas velocity in the mixing zone. This test generated 56 hours on coal feed and 49 hours of gasification, which was shut down three times because of a buildup of deposits in the mixing zone. Operation during the first 2 days of testing were at 950°C (1742°F) and resulted in deposits preventing solids circulation within 9 hours of entering gasification. The longest test period

of approximately 34 hours in gasification was achieved by dropping the operating temperatures 50°C (122°F) before another deposit forced a system shutdown. Table 8 shows all the average operating conditions from this test period.

TRDU Test P061 was a gasification test operated during the week of March 7–11, 1999, to test the ability of a transport reactor to gasify a near-term opportunity fuel such as petroleum coke. The Hunt Oil Refinery in Tuscaloosa, Alabama, was selected as the source because of its location near

Table 8. TRDU Tests P060 and P061 Operating Conditions

Parameter	P060	P061	P061
Condition	Air-blown gasification	Air-blown gasification	Enriched air gasification
Coal	SUFCo	Petcoke	Petcoke
Moisture Content, %	9.5	0.9	0.9
Pressure, psig	120	120	120
Steam:Coal Ratio, lb/lb coal	0.24	0.32	0.14
Air:Coal Ratio, lb/lb coal	2.9	2.8	NA
Ca:S Ratio, mole, sorbent only	2	1	1
Coal and Sorbent Feed Rate, lb/hr	272	335	520
J-Leg Zone, °C, avg.	820	873	984
Mixing Zone, °C, avg.	890	973	1088
Riser, °C, avg.	896	920	1017
Standpipe, °C, avg.	817	855	937
Dipleg, °C, avg.	673	668	714
TRDU Outlet, °C, avg.	849	869	980
Carbon Conversion, %	87	73	65
Carbon in Bed, %, standpipe	8–19	60	92.5
Riser Velocity, ft/sec	40–45	37.1	38.5
Standpipe Velocity, ft/sec	0.38–0.45	0.37	0.44
Calc. Circulation Rate, lb/hr	3000–4000	2565	2600
HHV of Fuel Gas, actual, Btu/scf	50–55	32	66
HHV of Fuel Gas, cor. Btu/scf	85–90	52	124
Duration, hr	49	24	26
Date (1998)	3/01–3/05	3/07–3/08	3/9–3/11

the PSDF in Wilsonville, Alabama. This fuel was tested under both air-blown and oxygen-enriched air-blown operation in the TRDU.

These operating conditions were interrupted twice because of solids plugging in the disengager solids drain back into the standpipe. The average operating conditions from these test conditions are also shown in Table 6. A small deposit in the burner gas entrance was found at the end of the test. This deposit was attributed to the low entrance velocity in the vicinity of the burner throat and the higher operating temperatures achieved with higher oxygen concentrations and less inert nitrogen associated with the air.

Table 9 shows the bulk chemical composition of SUFCo and petroleum coke steady-state solid samples obtained from the TRDU during the time period that these deposits formed. The SUFCo coal standpipe sample was approximately 200 μm in size, while the dipleg sample averaged 52 μm in size, and the filter ash was 15 μm in average size. The petroleum coke standpipe sample was approximately 500 μm in size and increasing throughout the test, while the dipleg material averaged 38 μm in size, and the filter ash averaged 9 μm .

4.2.2 TRDU Gasification Test P062

TRDU Test P062 was a gasification test operated on July 13 and 15, 1999, using a bituminous coal from the Calumet Mine in Alabama, and Longview Limestone from Alabama, which were selected because of their close proximity to the PSDF facility in Wilsonville. This test generated only 25 hours of coal feed and 10.5 hours of operation in coal gasification mode with the system gases and fly ash passing through the filter vessel during the whole test campaign. These tests were terminated early because of deposition and char agglomeration problems in the mixing zone with the Calumet fuel. Two instances of solids hangup in the disengager cyclone were also encountered during the heatup in combustion mode on Alabama bituminous coal. In both cases, the blockage cleared itself after coal feed was stopped and gas flow to the TRDU was reduced. The two tests with coal were very short because of the rapid buildup of deposits in the mixing zone. Compounding the operating problems was the buildup of char agglomerates in the mixing zone because of the higher-than-expected swelling properties of the bituminous coal. The operating temperature was approximately 1000°C (1832°F) in the mixing zone, but quickly dropped as deposit material covered the thermocouples. Coal feed was approximately 278 lb/hr with an air:coal ratio of 3.0 lb/lb coal and a steam:coal ratio of 0.27 lb/lb coal. Since the tests were so short, little satisfactory steady-state data were obtained.

4.2.3 TRDU Gasification Test P063

Another TRDU test campaign (TRDU Test P063) was conducted the week of August 29 – September 2, 1999, that generated approximately 90 hours of coal feed and slightly over 80 hours of gasification including 4 hours of enriched air gasification testing on the design Illinois No. 6 coal. Tests were conducted to examine the effects of air and steam distributions in the mixing zone, circulation rate, air/fuel and steam/fuel ratios on product gas heating value, and carbon conversion. Both air-blown and oxygen-enriched air-blown gasification tests were conducted during this test campaign. The range of average operating conditions obtained under the various test conditions of

Table 9. XRF Chemical Composition of TRDU Samples, Tests P060 and P061

Element	SUFCo	SUFCo	SUFCo	SUFCo	SUFCo	Petcoke	Petcoke	Petcoke	Petcoke	Petcoke
	Ash w/PRD	Deposit	Standpipe	Dipleg	Filter	Ash /w PRD	Deposit	Standpipe	Dipleg	Filter
Si	23.3	87.6	76.2	46.4	21.4	3.1	28.9	8.8	2.4	3.3
Al	6.9	0.6	2.8	5.3	7.5	0.2	1.4	0.9	0.3	0.8
Fe	6.3	1.3	2.3	3.4	5.9	0.6	2.3	2.4	0.3	0.8
Ti	0.7	0.2	0.3	0.4	0.9	0.1	0.2	0.3	0.1	0.1
P	0.1	0.1	0.1	0.1	0.2	0.0	0.0	0.0	0.0	0.0
Ca	33.6	5.5	9.6	24.3	36.5	44.3	40.7	27.5	46.5	46.4
Mg	13.5	3.2	4.1	12.8	12.8	22.9	19.8	12.3	21.3	24.0
Na	4.5	1.4	3.2	4.4	4.3	0.0	0.0	0.0	0.0	0.0
K	0.3	0.2	0.2	0.2	0.3	0.2	0.1	0.6	0.2	0.2
S	10.7	0.0	1.2	3.8	10.2	27.5	5.7	41.4	27.7	21.9
Ni	0.0	0.0	0.0	0.0	0.0	0.2	0.1	1.3	0.2	0.4
V	0.0	0.0	0.0	0.0	0.0	0.9	0.6	4.6	0.8	2.2
Total	99.9	100.1	100	100.1	100	100	99.8	100.1	99.8	100.1

Table 10. TRDU Test P063 Operating Conditions

Parameter	P063	P063	P063
Conditions	Air-blown gasification	Air-blown gasification	Enriched air gasification
Coal	Illinois No. 6	Illinois No. 6	Illinois No. 6
Moisture Content, %	8.5	8.5	8.5
Pressure, psig	120	120	120
Steam:Coal Ratio, lb/lb coal	0.23–0.54	0.31	0.46
Air:Coal Ratio, lb/lb coal	3.02–4.37	3.02	NA
O ₂ :Coal Ratio, lb/lb coal	0.70–1.01	0.7	0.76
Ca:S Ratio, mole, sorbent only	2	2	2
Coal and Sorbent Feed Rate, lb/hr	234–355	302	344
J-Leg Zone, °C, avg.	837–923	875	927
Mixing Zone, °C, avg.	928–1023	964	1039
Riser, °C, avg.	851–916	872	918
Standpipe, °C, avg.	820–896	829	882
Dipleg, °C, avg.	320–590	484	420
TRDU Outlet, °C, avg.	827–891	842	883
Carbon Conversion, %	71.6	68	62
Carbon in Bed, wt%, standpipe	1–7	3	3
Riser Velocity, ft/sec	26.1–38.2	27.7	33.8
Standpipe Velocity, ft/sec	0.25–0.31	0.25	0.26
Circulation Rate, lb/hr	2830–3800	3300	2575
HHV of Fuel Gas, actual, Btu/scf	40–56	56	64
HHV of Fuel Gas, corrected, Btu/scf	63–93	93	99
Duration, hr	80	4	4

reactor velocity, air:coal ratios, and circulation rates are shown in Table 10. Average operating conditions for the best air-blown test and the enriched air test are also shown in Table 10. Table 11 summarizes the bulk chemical composition of TRDU samples collected close to the time the mixing zone deposit was formed when the gasifier was being transitioned to the next highest oxygen enrichment condition. The hot-gas filter system was online for all but 4 hours of operation. These tests, especially the oxygen-enriched tests, indicated that significant modification of the TRDU to enhance circulation rate would be necessary before proceeding with plans to full oxygen-blown gasification tests on a transport reactor gasifier.

4.3 TRDU Modifications for Oxygen-Blown Transport Reactor Gasification Testing

4.3.1 Initial TRDU Modifications

After Test P063, the TRDU underwent a substantial modification in order to make improvements that would allow full oxygen-blown operation to occur. These modifications occurred over the period from the first quarter of 2000 until the first quarter of 2001.

Table 11. XRF Chemical Composition of TRDU Samples, Test P063

Element, wt%	Illinois No. 6	Illinois No. 6	Illinois No. 6 Standpipe	Illinois No. 6 Dipleg	Illinois No. 6 Filter
	Coal Ash w/ 17 wt% PRD	Mix Zone Deposit			
Si	23.0	26.2	19.6	17.2	14.8
Al	10.3	2.6	4.4	5.4	5.9
Fe	11.1	8.5	8.4	9.7	6.3
Ti	0.5	0.2	0.3	0.3	0.4
P	0.1	0.1	0.2	0.1	0.1
Ca	31.5	33.8	35.1	34.8	37.7
Mg	13.2	18.7	18.6	16.8	17.7
Na	0.4	0.0	0.0	0.0	0.0
K	1.5	0.5	1.0	1.1	0.9
S	8.5	9.3	12.3	14.7	14.3
Total	100.1	99.9	99.9	100.1	100.1

Specific needs for an oxygen-blown transport reactor were identified as:

1. Need for increased solids circulation: higher solids circulation dissipates the higher heat release generated by oxygen-blown operation.
2. A need for significantly higher steam flow rates and the need to mix the steam with the oxygen before it enters the gasifier and can contact circulating carbon in the bed material.
3. The need for a better gas seal for the dipleg solids return into the standpipe.
4. A coal feed nozzle that would inject the coal upward into the riser as compared to the original coal feed nozzle which would inject the coal feed at the top of mixing zone in a downward direction. There was concern that the downward-oriented coal feed nozzle was allowing fresh coal to back-mix into the partial oxidation region of the mixing zone, thereby combusting the fuel volatile matter instead of it just being volatilized and cracked in to the fuel gas.

Dr. Knowlton's recommendations are summarized as installing an L-valve as the standpipe loop seal design over the original J-leg or alternate Y-leg design. The J-leg had been shown to present some solids recirculation restrictions as actual solids recirculation rates were only about 1/10 of the design recirculation rate expected. This low solids recirculation rate is due to wall effects and to the large momentum change required to get the solids to turn the corner from the standpipe into the J-leg and flow back uphill through the J-leg to the mixing zone. The alternate choices were between an L-valve or a Y-leg with each loop seal having some advantages and disadvantages unique to that particular loop seal. A major advantage for the Y-leg included the minimization of aeration gas required to return the solids to the bottom of the mixing zone. A disadvantage of the Y-leg (at least in the TRDU design with a fairly wide spacing between the riser and the standpipe) would be fairly large bubbles accumulating on the upper side of the inclined Y-leg, and these bubbles could potentially inhibit solids circulation as the large bubbles would enter the vertical standpipe. Another disadvantage for the EERC transport reactor was the Y-leg would have to pass

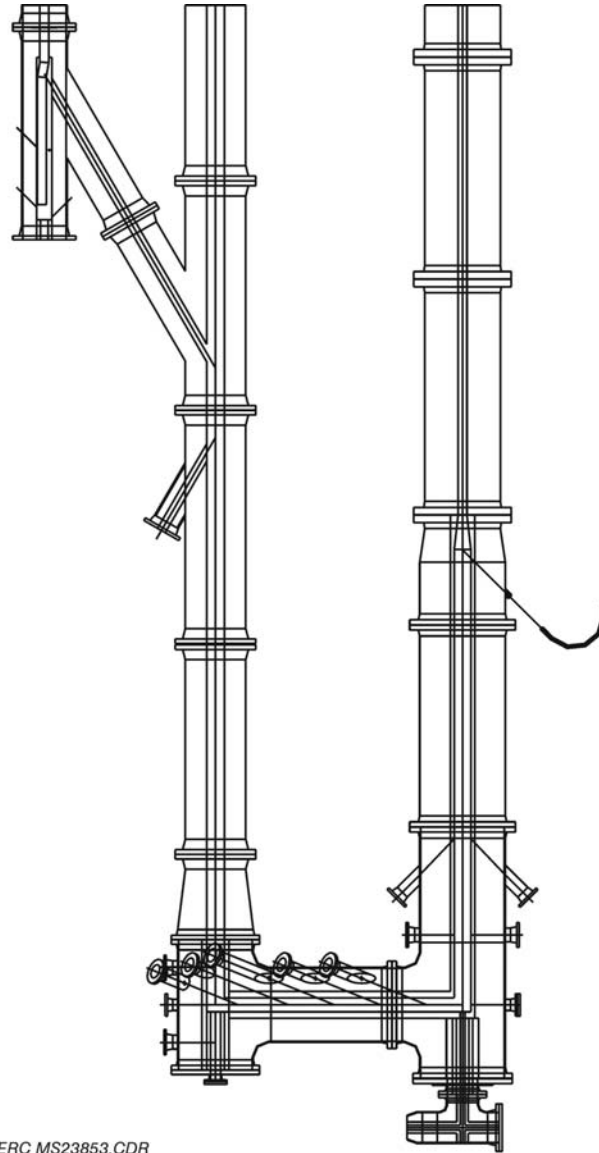
through an area currently occupied by a major structural beam requiring major structural modification of the building support structure.

The other potential loop seal design was an L-valve that would also increase solid circulation rates over the J-leg and reduce the amount of inert or recycled syngas injected to recirculate the solids back to the mixing zone. However, this design would probably still require significantly more gas being injected in this area to move the solids across the horizontal L-valve than would be required for a Y-leg loop seal. Another advantage would be the increased length on the mixing zone would add some additional residence time in the mixing zone, and construction would be simplified because of the use of conventional pipe tees instead of utilizing a specially constructed pipe-Y in the bottom of the mixing zone.

Other recommendations were to increase the standpipe diameter and dipleg diameter to minimize the effect of bubble size and friction effects with the wall on solids flowing down either the standpipe or dipleg. Another recommendation was to add a loop seal to the bottom of the dipleg so that the standpipe inventory could be operated independently of the level where the dipleg solids reentered the standpipe. Without a loop seal in the bottom of the dipleg, the standpipe inventory always had to operate at a level greater than the dipleg solids return height or a significant amount of gas would bypass up the dipleg, thereby severely spoiling the performance of the primary cyclone.

In addition, a simple atmospheric-pressure cold-flow system was constructed to determine the effects of different loop seals and mixing zone sizes on the amount of solids circulation, backmixing, and residence time in the mixing zone. The sizes investigated were approximately 87% of TRDU-scale equivalent mixing zone dimensions. Since the unit can only operate under atmospheric pressure and the same starting bed material was used for the TRDU, the particle-to-gas density ratio was not matched in these tests. Air and nitrogen flows were adjusted to match the relative distribution in the TRDU and to match the desired riser velocity (at atmospheric pressure and temperature). The mixing zone test sizes were selected to be equivalent to the original mixing zone diameter, a diameter 2 inches larger and another 4 inches larger than the original mixing zone. Cold-flow tests utilizing a 25-pound batch of red-colored FCC catalyst support material as a tracer were completed for each mixing zone configuration. The colored sand residence time tests were conducted using both burner air only and a combination of burner air and combustion air. From these tests, it was possible to measure residence time distributions passing through the mixing zone. These residence times were measured at 50, 65, and 75 seconds for the amount of time that the colored FCC material remained in the mixing zone. Based on these results, a diameter matching the medium diameter was selected for the TRDU design, and this residence time distribution was utilized in the calculation of the steam/carbon kinetics shown in Appendix A.

Based on the above recommendations and the results of the cold-flow testing, the L-leg loop seal design was selected as the best design given the building and budget constraints for the project. Figures 4 through 6 show the as-designed L-valve modification for the TRDU. The modifications consisted of constructing three new sections as shown in Figure 4. One section would be for the L-valve loop seal (Figure 5), and the other two sections would replace the J-leg elbow (also shown in Figure 6) and the bottom section of the TRDU mixing zone. In addition, as shown in Figure 6,



EERC MS23853.CDR

Figure 4. Schematic of TRDU modifications made for oxygen-blown operation.

Section Q on the bottom of the dipleg was modified to allow for a seal pot to be installed as the loop seal on the bottom dipleg. Other modifications included enlarging the standpipe and dipleg diameters by boring out the refractory in order to minimize wall effects on the flowing solids. The standpipe diameter was increased by 2.5 inches, and the dipleg diameter was increased by 0.625 inches, respectively. In addition to reducing wall effects, this increased diameter should reduce the bridging of the circulating solids across the bottom of the cyclone cones. This interrupts circulation of the bed material, and extended interruptions can lead to loss of bed material and eventually increased bed agglomeration and deposition. Instrumentation to measure and control the flow of oxygen and steam to the burner and mix zone ports was also added to allow the TRDU to be operated under full oxygen-blown conditions.

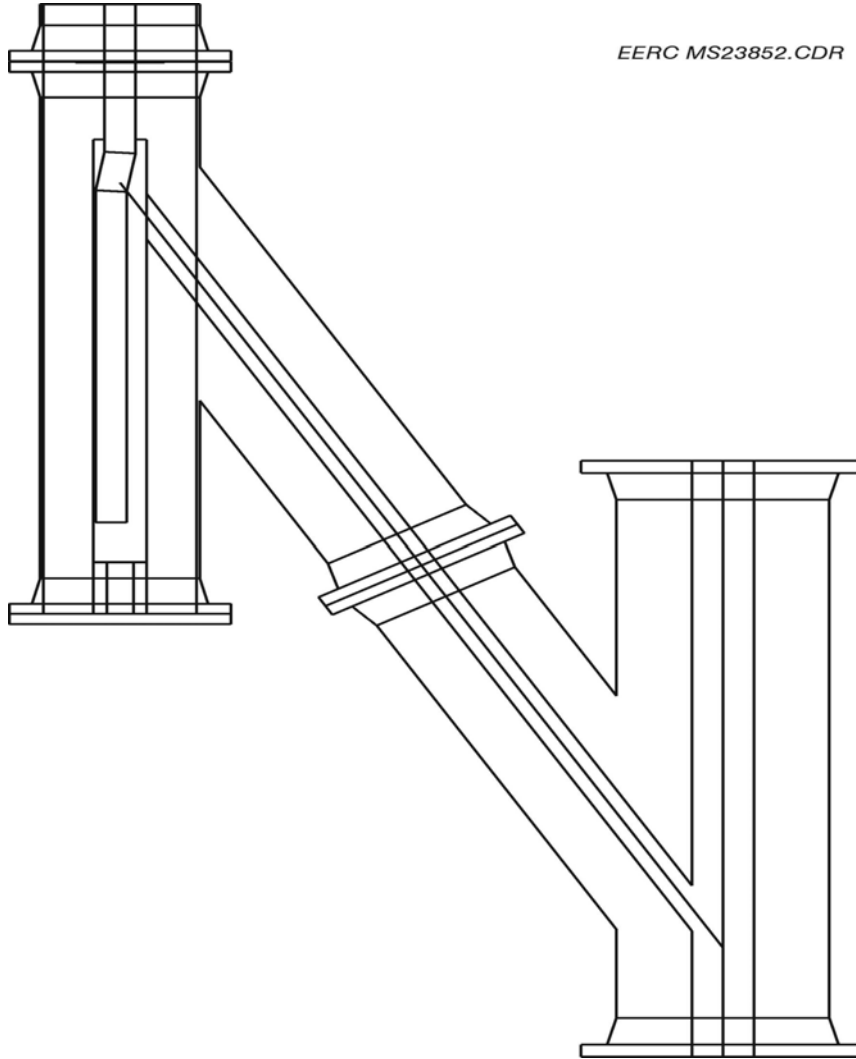


Figure 5. Drawing of L-valve and lower mixing zone modifications.

4.3.2 Additional TRDU Modifications

After Test P069, a thermal oxidizer was installed after the hot-gas filter system to combust the hot fuel gas from the TRDU and avoid the gas quench system with its numerous problematic and costly issues associated with the water scrubbers, circulating pumps, heat exchanger plugging, and wastewater disposal issues. Figure 7 shows a schematic of the thermal oxidizer built for the TRDU system. This thermal oxidizer was designed to handle all of the gas flow from the TRDU to combust this fuel gas between 1800° to 2000°F (982° to 1093°F) with a 2-second residence time. The thermal oxidizer is started and heated up on natural gas delivered down a central burner nozzle along with a substoichiometric amount of primary combustion air. Secondary air is swirled in around the primary burner and provides all of the combustion air to finish the combustion of the natural gas and all of the fuel gas from the TRDU. The hot fuel gas from the TRDU is then swirled around the

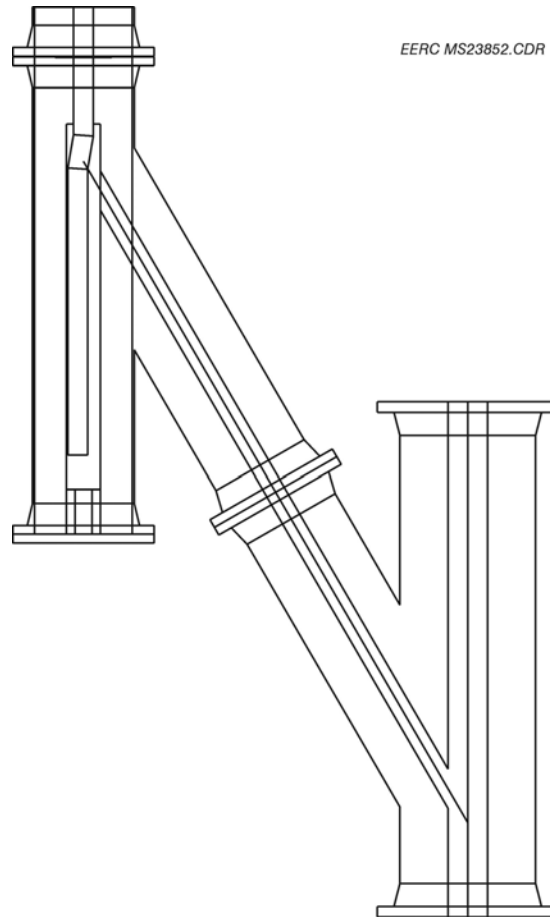


Figure 6. Drawing of seal pot loop seal on bottom of the dipleg.

outside of the secondary air to complete the combustion process. The flame safety system is set to add supplemental natural gas in order to maintain thermal oxidizer temperature; however, most tests have shown that the thermal oxidizer can be operated without any supplemental fuel. All extra air from the blower is added to the stack flue gas to cool the gas before it is discharged out the stack. No baghouse or particulate collection device is included downstream of the thermal oxidizer. Since there is no backup particulate control, the thermal oxidizer was installed such that the fuel gas flow can be diverted from the thermal oxidizer and sent to the quench system should a major filter failure result in a high dust loading to the thermal oxidizer. Collection of all the gaseous flow rates and gas emissions data including moisture concentration allows a material balance around the thermal oxidizer to be completed. This allows another measure of carbon conversion, sulfur retention, and fuel gas heating value to be calculated from the data collected.

4.4 Oxygen-Blown Results Utilizing the TRDU L-Valve Modification

In total, ten test campaigns have been conducted under enriched air or full oxygen-blown conditions. During these tests, 1515 hours of coal feed with 660 hours of air-blown gasification and

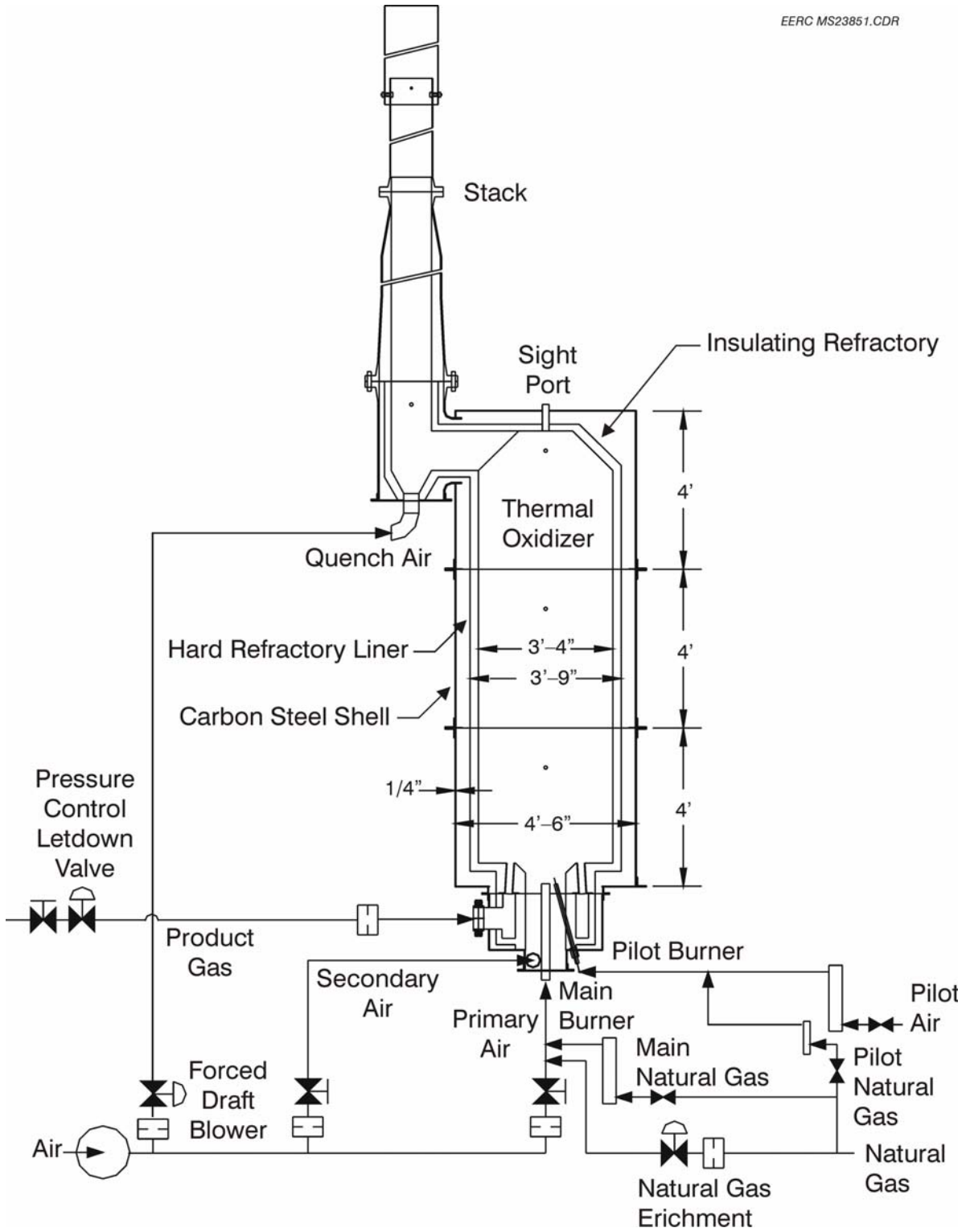


Figure 7. Schematic of the thermal oxidizer added to combust the TRDU fuel gas.

720 hours of enriched air or oxygen-blown gasification were completed. During these tests, approximately 366 hours of operation with Wyodak, 91 hours of operation with Navajo subbituminous coal, 135 hours of operation on Illinois No. 6, 108 hours on SUFCo, 110 hours on Prater Creek, 48 hours on Calumet, and 134 hours on a Pittsburgh No. 8 bituminous coal were completed. In addition, 287 hours of operation on low-rank coals such as North Dakota lignite, Australian brown coal, and a 90:10 wt% mixture of lignite and wood waste were completed. Also included in these test campaigns was 50 hours of gasification on a petroleum coke from the Hunt Oil Refinery. An additional 73 hours of gasification on a high ash Indian coal was completed.

Detailed operating and material balance information for all of the steady-state air and oxygen-blown tests is given in Appendix A. Elemental (H, C, N, S, O) material balances based on measured inputs are also given in Appendix A. This information was also utilized with a simple model based on thermogravimetric analysis (TGA) kinetics to estimate the amount of steam/carbon gasification and partial oxidation that is occurring with the circulating char (3–5). In general, these estimates showed that while partial oxidation of the recirculating carbon contributed significantly to the conversion of carbon, the conversion due to the steam–carbon reaction was small to moderate (depending on the char kinetics and the residence time distribution utilized). This implies that a large portion of the fuel gas heating value is being derived from the devolatilization and cracking of the fuel volatile matter.

4.4.1 TRDU Oxygen-Blown Shakedown Tests P066 and P067

4.4.1.1 TRDU Gasification Test P066

Test P066 was a shakedown test using the new TRDU L-valve loop seal and was accomplished during the weeks of January 23 – February 1, 2001, and that generated 125 hours and 137 hours of coal feed, respectively, with 109 and 121 hours of operation in coal gasification mode, respectively. These tests were to shake down the TRDU oxygen-blown modifications. These tests were completed at a full system pressure of 120 psig and were conducted while the steam flow was split between the L-valve nozzles and the burner and mix zone steam/oxygen ports. This resulted in fluctuating steam flows that were hard to control. This prevented the TRDU from operating in full oxygen-blown mode. Oxygen/air flow ratios up to 35% oxygen were achieved, but attempts at full oxygen-blown operation resulted in mixing zone deposition and agglomeration. A coal feed nozzle that pneumatically transported the coal directly into the bottom of the riser was tested; however, this testing resulted in extremely high tar concentrations in the fuel gas and difficulty in keeping the hot gas filter baseline differential pressure stable. The coal feed nozzle was switched back to its original downward configuration after 3 days of operation.

4.4.1.2 TRDU Gasification Test P067

Test P067, the second shakedown test using the new TRDU L-valve loop seal, was accomplished during the weeks of May 13–22, 2001. This test utilized Navajo subbituminous coal, Wyodak subbituminous coal, and Illinois No. 6 bituminous coal. This test generated 137 hours of coal feed with 59 hours of air-blown gasification and 62 hours of enriched and oxygen-blown gasification (121 hours in gasification total). This test was run at reduced pressure in order to utilize University of North Dakota (UND) steam as a second source of steam to the TRDU. This steam,

which is only available at 130 psig, required the TRDU to operate at 100 psig. Table 12 shows some of the operating conditions from selected test periods on these three coals. Table 13 shows the actual and corrected product gas compositions for these same test periods.

4.4.2 TRDU Gasification Test P068

This test campaign was conducted as part of a test program conducted with the North Dakota Industrial Commission's Lignite Research Council to evaluate North Dakota lignite in a transport reactor (6). This test was the first to successfully operate the transport reactor in full oxygen-blown mode without bed material agglomeration and deposition problems. Results from these tests are shown in Appendix A along with the data from all of the other oxygen-blown tests. The TRDU was operated at average temperatures ranging from 792° to 828°C (1458° to 1522°F) at various air/fuel ratios and reactor velocities. Table 14 summarizes the range of operational performance for the TRDU during these test periods. Table 15 summarizes the optimum operating conditions achieved with each lignite. In general, similar actual and corrected fuel gas heating values were achieved with all three lignites. The actual dry product gas produced was 4.7% to 7.4% CO, 12.7% to 20.8% H₂, 20.8% to 29.7% CO₂, 1.9% to 3.1% CH₄, and 0.20% to 0.35% ethane with the balance being N₂ and other trace constituents. The moisture in the fuel gas exiting the transport reactor ranged from 45.9% to 55.7% under oxygen-blown conditions. Coal/sorbent feed rates ranged from 413 to 586 lb/hr, and the gasifier pressure averaged 100 psig. Calculated recirculation rates ranged from 950 to 7650 lb/hr.

The recirculating bed material particle size for Test P068 was approximately 150 to 300 μm while on the Montana limestone. The circulating bed material with the Falkirk lignite was approximately 180 μm and increased to approximately 400 μm while on the Freedom coal with the Montana limestone. After restarting the TRDU on fresh sand for the high-sodium Freedom lignite test with the Plum Run dolomite, the bed material particle size remained relatively constant at 150 μm. The particle-size distribution for the filter ash for all coals was approximately 15 μm but ranged between 8 μm up to almost 20 μm. Comparisons of particle-size distributions for selected standpipe and filter samples collected show there was no significant difference in particle-size distributions between air-blown and oxygen-blown operation.

Correction of the raw product gas stream is necessary because of the high level of dilution caused by the nitrogen purges in the system and by the high heat losses as a percentage of the coal feed experienced by a pilot-scale system. These corrections assume that the purges would either be small enough to be inconsequential, or when significant amounts of purge gas are required, a compressor would recycle syngas instead of injecting nitrogen. Heat losses were corrected from approximately 15% of the coal feed heat input to approximately 0.25% of the coal feed heat input. Comparing the corrected product gas compositions, the air-blown fuel gas composition would be 15%–17% H₂, 9%–12% CO, 2.0%–3.0% CH₄, and 15%–17% CO₂, as compared with a corrected fuel gas composition of 35%–39% H₂, 13%–14% CO, 4.5%–6.0% CH₄, and 38%–41% CO₂. The high hydrogen and carbon dioxide concentrations under oxygen-blown conditions are the result of the water-gas shift reaction (shown below) being driven to form the products on the right-hand side of the equation by the high steam partial pressure in the gasifier product gas stream:

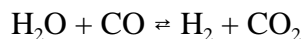


Table 12. TRDU Gasification Efficiency for Test P067

Test	Navajo-12	Navajo-14	Wyodak-16	Wyodak-17	Illinois No. 6-22	Illinois No. 6-23
Gasifier Temp., °C	847	899	816	825	858	884
Coal/Sorbent Feed Rate, lb/hr	425	450	373	390	422	428
Air Flow, lb/hr	918	271	915	268	870	274
O ₂ Flow, lb/hr	0	145	0	145	0	152
Steam Flow, lb/hr	336	335	332	369	408	504
Air:Coal Ratio, lb/lb	2.4	0.67	2.56	0.72	2.48	0.77
Steam:Coal Ratio, lb/lb	0.88	0.83	0.93	0.99	1.16	1.42
O ₂ :Coal Ratio, lb/lb	0.56	0.51	0.59	0.55	0.58	0.61
Recirculation Rate, lb/hr	4362	14,634	3612	3145	3844	4268
TRDU Throughout, lb/hr-ft ²	5726	6063	5360	5605	5243	5318
TRDU Throughout, MMBtu/hr-ft ²	50.4	53.4	52.3	54.6	59.3	60.1
TRDU Riser Velocity, ft/s	54.3	44.2	56.7	46.6	54.5	51.1
Carbon Conversion						
Solid Accountability	94.6	79.5	85	78.6	72.6	81.8
Gas Make	59.4	62.4	73.1	69.1	67.3	57.8

Table 13. Corrected TRDU Product Gas Compositions for Test P067

Test	Navajo-12	Navajo-14	Wyodak-16	Wyodak-17	Illinois No. 6-22	Illinois No. 6-23
Product Gas Composition, vol%						
H ₂	5.48	11.19	7.63	16.79	3.55	11.7
CO	2.74	6.83	3.95	7.03	2.56	6.35
CH ₄	1.59	3.61	1.68	2.94	1.54	3.94
CO ₂	13.34	19.32	13.86	19.44	13.56	20.67
N ₂	75.59	58.7	72.71	53.77	78.17	56.07
Total	99.12	99.65	99.83	99.97	99.38	98.73
Heating Value, Btu/scf	47	95	55	107	35	98
% N ₂ in Feed	26.4	41.9	28.3	49.8	26.1	43.6
N ₂ -Free Heating Value, Btu/scf	59	126	69	152	45	140
Product Gas, vol%		Adjusted for 450,000 Btu/hr heat loss and N ₂ purge free				
H ₂	8.6	20.7	11.6	31.5	5.4	22.2
CO	4.3	12.7	6	13.2	3.9	12.1
CH ₄	3.1	6.7	2.5	5.5	2.4	7.5
CO ₂	16.3	28.3	17.2	30.4	16.9	33
N ₂	67.7	31.6	62.7	19.5	71.4	25.3
Total	100	100	100	100	100	100
Heating Value, Btu/scf	73	176	83	200	54	187

Table 14. TRDU Range of Oxygen-Blown Operating Conditions for North Dakota Lignites

Parameter	P068	P068	P068
Conditions	Gasification	Gasification	Gasification
Coal	Center	Falkirk	Freedom
Moisture Content, %	35	36.2	28.3–33.8
Pressure, psig	100	100	85–100
Steam:Coal Ratio	0.80–1.01	0.79–0.95	1.05–1.43
O ₂ :Coal Ratio	0.41–0.51	0.40–0.48	0.49–0.61
Ca:S Ratio, mole (total including ash)	1.61	1.52	1.2–3.5
Coal and Sorbent Feed Rate, lb/hr	440–567	487–586	413–531
Avg. Mixing Zone Temp, °C , avg.	808–828	796–811	792–816
HHV of Fuel Gas, act., Btu/scf	79–128	112–121	90–118
HHV of Fuel Gas, cor., Btu/scf	220–239	211–236	214–232
Conversion, %	79–90	80–87	64–90
Carbon in Bed, %, standpipe	6–26	6–25	5–37
Riser Velocity, ft/s	41.6–45.0	50–53	48–51
Standpipe Velocity, ft/s	0.35	0.35	0.35
Circulation Rate, lb/hr	3250–7650	4000–5000	950–2550
Duration, hr	77	46	67

This high hydrogen and carbon dioxide product gas stream would make an excellent gas stream for hydrogen separation and for CO₂ separation and possible sequestration under a Vision 21 project. Table 16 shows the ash analysis from the lignite oxygen-blown gasification tests. These data suggest that even though the Freedom lignite was high in sodium, the bed material did not appear to be accumulating a lot of sodium. This presumably is due to a large majority of the sodium in the Freedom lignite being organically associated, which should result in the formation of fine sodium aerosol fume that will pass through the cyclones and condense on the filter ash and carbon.

4.4.3 TRDU Gasification Test P069

Another test campaign (P069) was conducted during the week of October 8–15, 2001. During this week, approximately 150 hours of coal feed and 143 hours of gasification, respectively, were achieved, with the system gases and fly ash passing through the filter vessel during the whole test campaign. Test P069 was terminated early because of a hot spot on one of the L-valve nozzles. This hot spot was the result of erosion of the soft insulating refractory around one of the downward-oriented L-valve nozzles. Table 17 shows selected operating data from this test. Table 18 shows a comparison of the fuel gas compositions for both air-blown and oxygen-blown operation. Appendix A shows all of the steady-state operating periods obtained for this test, including material balances.

Table 15. TRDU Optimum Oxygen-Blown Operating Conditions for North Dakota Lignites

Parameter	P068	P068	P068
Conditions	Gasification	Gasification	Gasification
Coal	Center-6	Falkirk-2	High-Na Freedom-3
Pressure, psig	100	100	85
Steam:Coal Ratio	1.01	0.92	1.05
O ₂ :Coal Ratio	0.5	0.47	0.49
% Sorbent in feed, wt%	8	10	20
Coal and Sorbent Feed Rate, lb/hr	457	502	531
Avg. Mixing Zone Temp., °C	812	8.11638e+14	8.09580713e+14
Avg. L-Valve Temp., °C	700		
Avg Riser Temp., °C	728		
Avg Standpipe Temp., °C	748		
Avg. Dipleg Temp., °C	440		
Conversion, %	85	85.3	81.5
Product Gas HHV (act.), Btu/scf	234	233	232
Prodcut Gas HHV (cor.), Btu/scf	–	–	–
Carbon in Bed, %, standpipe	10.7	12.2	10.4
Riser Velocity, ft/s	42.5	42.4	48.8
Standpipe Velocity, ft/s	0.35	0.35	0.35
Circulation Rate, lb/hr	5200	4000	947

4.4.4 TRDU Gasification Test P070

Test P070 was only scheduled for a week of operation from April 15 to 20, 2002. During this test, 118 hours of coal feed and 110 hours of gasification were completed, including 36 hours in air-blown gasification and 74 hours in enriched air and full oxygen-blown operation. This test utilized SUFCo and Illinois No. 6 bituminous coal with most of the testing completed on the SUFCo coal. Operation was very steady, with the best air- and oxygen-blown results for both feedstocks shown in Tables 19 and 20. All of the data from the identified steady-state periods are also shown in Appendix A.

Table 16. XRFA Analysis of Oxygen-Blown TRDU Samples Generated from North Dakota Lignite

	6/18/01					6/21/01	
	Falkirk	Standpipe	Dipleg	Filter	Freedom	Standpipe	Filter
Si	24.9	22.9	33.4	29.9	25.3	16.4	10.3
Al	8.8	6.1	5.3	11.5	14.2	4.9	6.4
Fe	7.8	4.9	5.3	9	7.5	5.1	4.6
Ti	0.5	0.4	0.4	0.6	0.6	0.3	0.4
P	0.1	0.1	0.1	0.1	0.3	0.1	0.2
Ca	34.8	56.5	47.9	35.8	20.8	42.7	47.1
Mg	4	3.9	2.6	6.2	8	25.0	26.5
Na	3.5	2.2	3.2	3.6	11.8	3.2	2.8
K	1.8	1.6	1.3	2	1.4	0.9	0.5
S	13.8	1.4	0.5	1.3	10.1	1.4	1.2
Total	100	100	100	100	100	100	100

Table 17. Corrected TRDU Product Gas Compositions for P069

Test	Wyodak	Wyodak
Product Gas Composition, vol%	Air	Oxygen
H ₂	10.0	19.1
CO	5.7	11.0
CH ₄	1.7	3.9
CO ₂	13.5	24.2
N ₂	79.0	45.4
Total	109.8	103.6
Heating Value, Btu/scf	68	137
% N ₂ in Feed	28.6	64.1
N ₂ -Free Heating Value, Btu/scf	76	192
Product Gas, vol%	Corrected for Heat Losses and N ₂ Purge Free	
H ₂	13.7	35.2
CO	7.7	20.4
CH ₄	2.3	7.2
CO ₂	14.3	35.6
N ₂	62.0	1.7
Total	100	100
Heating Value, Btu/scf	93	253

Table 18. TRDU Gasification Efficiency for Test P069

Test	Wyodak	Wyodak
Oxidant	Air	Oxygen
Gasifier Temp., °C	823	892
Coal/Sorbent Feed Rate, lb/hr	476	406
Air Flow, lb/hr	988	0
O ₂ Flow, lb/hr	0	239
Steam Flow, lb/hr	269	364
Air:Coal Ratio, lb/lb	2.16	0
Steam:Coal Ratio, lb/lb	0.58	0.93
O ₂ :Coal Ratio, lb/lb	0.50	0.61
Recirculation Rate, lb/hr	1530	3665
TRDU Throughput, lb/hr-ft ²	6840	5835
TRDU Throughput, MMBtu/hr-ft ²	61.3	52.6
TRDU Riser Velocity, ft/s	54.8	44.0
Carbon Conversion		
Solid Accountability	79.4	95.5

Table 19. Corrected TRDU Product Gas Compositions for P070

Test	SUFCo	SUFCo	Illinois No. 6
Product Gas Composition, vol%	Air	Oxygen	Oxygen
H ₂	10.7	15.8	14.7
CO	4.6	7.4	6.8
CH ₄	1.9	3.8	3.2
CO ₂	13.2	22.3	20.1
N ₂	70.8	53.7	57.1
Total	101.2	103.0	101.9
Heating Value, Btu/scf	68	114	102
% N ₂ in Dry Feed Cases	26.6	65.6	55.8
N ₂ -Free Heating Value, Btu/scf	88	189	162
Product Gas, vol%	Adjusted for 450,000 Btu/hr Heat Loss and N ₂ Purge Free		
H ₂	17.4	35.3	37.0
CO	7.5	16.5	17.1
CH ₄	3.1	8.5	8.0
CO ₂	16.2	37.1	34.5
N ₂	55.8	2.6	3.4
Total	100	100	100
Heating Value, Btu/scf	112	254	257

Table 20. TRDU Gasification Efficiency for P070

Test	SUFCo	SUFCo	Illinois No. 6
Oxidant	Air	Oxygen	Oxygen
Gasifier Temp., °C	880	900	972
Coal/Sorbent Feed Rate, lb/hr	412	296	295
Air Flow, lb/hr	906	0	57
O ₂ Flow, lb/hr	0	210	222
Steam Flow, lb/hr	287	454	483
Air:Coal Ratio, lb/lb	2.29	0	0.23
Steam:Coal Ratio, lb/lb	0.73	1.59	1.89
O ₂ :Coal Ratio, lb/lb	0.64	0.90	1.19
Recirculation Rate, lb/hr	4120	6740	4615
TRDU Throughout, lb/hr-ft ²	5920	4253	3665
TRDU Throughout, MMBtu/hr-ft ²	68.6	49.3	41.4
TRDU Riser Velocity, ft/s	57.1	57.2	61.4
Carbon Conversion			
Solid Accountability	76	83.0	81.0
Gas Make	77	56	57.4

4.4.5 TRDU Gasification Test P071

Test campaign P071 was run during the June 9–16, 2002, time period. This test had 107 hours of coal feed and 98 hours of gasification, including 18 hours of air-blown gasification and 80 hours of enriched air- or full oxygen-blown gasification. This test utilized Tuscaloosa petroleum coke and Prater Creek bituminous coal as feedstocks, with the petcoke sized to –30 mesh and the Prater Creek bituminous coal sized to the standard –10-mesh particle size. This test was ended prematurely when another hot spot on the riser vessel wall was detected. This hot spot was the result of too short of a ceramic plug being inserted into a downward pointing secondary air nozzle in the riser. Over the years of testing at the EERC, the high-velocity bed material was able to start eroding the soft insulating refractory behind the hard face refractory, creating a significant hollow pocket that would fill with hot-bed material. After this test, the EERC performed a through inspection of all the TRDU sections, paying particular attention to all nozzle penetrations. Any necessary refractory repairs were made, and either metal liner or ceramic plugs of the proper length were installed such that high-velocity bed material could not impact the softer insulating refractory. Tables 21 and 22 show the fuel gas compositions and the operating conditions achieved for these tests. All steady-state periods for this test campaign are given in Appendix A.

Table 21. Corrected TRDU Product Gas Compositions for Test P071

Test	Petcoke	Petcoke	Prater Creek
Product Gas Composition, vol%	Air	Oxygen	Oxygen
H ₂	6.7	18.8	14.2
CO	5.6	9.9	7.8
CH ₄	0.6	1.6	3.0
CO ₂	12.0	20.1	15.8
N ₂	74.0	50.2	61.4
H ₂ S, ppm	758	4798	1320
Total	98.8	100.6	102.3
Heating Value, Btu/scf	46	109	102
% N ₂ in Feed	29.3	62.0	60.1
N ₂ -Free Heating Value, Btu/scf	64	194	161
Product Gas, vol%	Adjusted for 450,000 Btu/hr Heat Loss and N ₂ Purge Free		
H ₂	11.0	35.9	37.5
CO	9.2	18.8	20.7
CH ₄	1.0	3.0	7.9
CO ₂	15.5	31.3	31.2
N ₂	63.3	11.0	2.7
Total	100	100	100
Heating Value, Btu/scf	76	207	269
Sulfur Retention %	88.6	49.9	30.0

Table 22. TRDU Gasification Efficiency for TRDU Test P071

Test	Petcoke	Petcoke	Prater Creek
Oxidant	Air	Oxygen	Oxygen
Gasifier Temp., °C	1020	965	980
Coal/Sorbent Feed Rate, lb/hr	289	241	329
Air Flow, lb/hr	1025	57	57
O ₂ Flow, lb/hr	0	216	235
Steam Flow, lb/hr	261	456	473
Air:Coal Ratio, lb/lb	4.73	0.32	0.19
Steam:Coal Ratio, lb/lb	1.2	2.52	1.52
O ₂ :Coal Ratio, lb/lb	1.10	0.27	0.80
Recirculation Rate, lb/hr	2260	2730	5350
Operation Pressure, psig	100	80	80
TRDU Riser Velocity, ft/s	53.7	64.2	67.4
Carbon Conversion			
Solid Accountability	78.2	77.5	74.1
Noncondensable Gas Make	58.1	71.3	51.7
Thermal Oxidizer	75.8	68.1	81.8

4.4.6 TRDU Gasification Test P072

Test P072 was conducted during the weeks October 7–9, 2002, and October 21–25, 2002, with an 11-day interruption caused by a gasket failure on the filter vessel tube sheet that seals the tube sheet between the filter vessel flanges. This resulted in fuel gas leaking directly into the gasification tower and forced a system shutdown. The run was restarted after the filter was cooled and the filter gasket replaced. This test operated for 155 hours on coal feed with 145 hours in gasification, including 20 hours in air-blown gasification and 125 hours in enriched air- or full oxygen-blown gasification. This test utilized Illinois No. 6, Alabama bituminous coal from the Calumet Mine and, for a short period, the high-swelling Pittsburgh No. 8 bituminous coal. Tables 23 and 24 show the corrected and actual fuel gas composition for oxygen-blown testing and the operation conditions for these selected tests on these fuels. All steady-state data from this test campaign are given in Appendix A.

4.4.7 TRDU Gasification Test P073

TRDU Test P073 was conducted from the April 22 to April 30, 2003, time period. This test generated 135 hours of coal feed with 120 hours of gasification, including 75 hours of air-blown gasification and 45 hours of oxygen-blown gasification. This test utilized a high-swelling Pittsburgh No. 8 bituminous coal from the Blacksville Mine exclusively. While this test generated a significant number of hours, it was also subject to a much higher number of significant fluctuations in solids circulation possibly because of the swelling properties of this coal. It was speculated that a layer of sticky coal could build up on the wall opposite where the coal is injected and then spall off in large enough agglomerates to cause the fluctuations in the circulation rates. Because of these fluctuations, the amount of true steady-state data appears to be limited. Tables 25 and 26 show the product gas composition and TRDU operating conditions and efficiency results for the high-swelling Pittsburgh No. 8 bituminous coal from the Blacksville Mine in West Virginia. Table 27 shows the XRFA analysis from bed material and filter ash samples generated during the gasification of the Blacksville coal. Again, the data from all the identified steady-state periods are given in Appendix A.

4.4.8 TRDU Gasification Test P074

TRDU Test P074 was conducted during the week of September 22 through September 28, 2003. This test generated 81 hours of coal feed with 65 hours of gasification data. Of this testing, 48 hours was in air-blown operation, and 17 hours was in oxygen-blown operation. The first part of this test attempted to test Australian brown coal from the Loy Yang Mine; however, steady-state operation was difficult to obtain since this original 60% moisture coal could only be air-dried to approximately 35%–40% moisture before testing was started. This fuel proved to be very difficult to feed, so this testing was discontinued. The test was then completed on a 90 wt% Falkirk, North Dakota, lignite and 10 wt% hog fuel wood waste feedstock. This testing represented the bulk (44 hours) of the good steady-state operating results obtained during this test campaign. Tables 28 and 29 show some results from these gasification tests on the lignite/wood mixture in both air-blown and oxygen-blown operation. Table 30 shows the ash chemistry from various bed material and filter vessel samples taken during the testing of both the Falkirk lignite and the Falkirk lignite–10 wt% wood mixture. This table shows that after 2 days of operation on the Falkirk–wood mixture, potential

Table 23. Corrected TRDU Product Gas Compositions for P072

Test	Illinois No. 6	Illinois No. 6	Calumet
Product Gas Composition, vol%	Oxygen	Oxygen	Oxygen
H ₂	12.6	11.1	16.8
CO	6.5	6.0	9.3
CH ₄	3.3	2.1	3.5
CO ₂	19.6	16.3	20.3
N ₂	59.5	64.1	48.7
H ₂ S, ppm	2461	3291	1593
Total	101.5	99.6	98.5
Heating Value, Btu/scf	95	76.0	119
% N ₂ in Noncondensable Feed	64.0	56.7	60.3
N ₂ -Free Heating Value, Btu/scf	176	120	219
Product Gas, vol%	Adjusted for 450,000 Btu/hr Heat Loss and N ₂ Purge Free		
H ₂	29.4	30.5	33.8
CO	15.2	16.6	18.7
CH ₄	7.8	5.6	7.0
CO ₂	33.1	31.4	31.1
N ₂	14.5	15.8	9.5
Total	100	100	100
Heating Value, Btu/scf	223	210	241
Sulfur Retention %	73.2	55.0	31.0

Table 24. TRDU Operating Conditions and Gasification Efficiency Results for P072

Test	Illinois No. 6	Illinois No. 6	Calumet
Oxidant	Oxygen	Oxygen	Oxygen
Gasifier Temp., °C	934	1016	987
Coal/Sorbent Feed Rate, lb/hr	334	294	351
Air Flow, lb/hr	57	57	054
O ₂ Flow, lb/hr	194	229	228
Steam Flow, lb/hr	416	438	434
Air:Coal Ratio, lb/lb	0.20	0.23	0.16
Steam:Coal Ratio, lb/lb	1.50	1.79	1.33
O ₂ :Coal Ratio, lb/lb	0.75	0.99	0.74
Recirculation Rate, lb/hr	4805	7315	6610
Operation Pressure, psig	85	85	85
TRDU Riser Velocity, ft/s	56	61.7	59.6
Carbon Conversion			
Solid Accountability	68.7	85.8	72.1
Noncondensable Gas Make	58.2	60.6	55.5
Thermal Oxidizer	74.3	82.2	74.3

Table 25. Corrected TRDU Product Gas Compositions for P073

Coal	- 10 mesh Blacksville	- 10 mesh Blacksville
Product Gas Composition, vol%	Air	Oxygen
H ₂	6.3	16.2
CO	4.7	9.5
CH ₄	1.8	4.1
CO ₂	12.8	23.1
N ₂	81.1	54.7
H ₂ S, ppm	892	2102
Total	106.7	107.6
Heating Value, Btu/scf	54	125
% N ₂ in Noncondensable Feed	29.6	53.1
N ₂ -Free Heating Value, Btu/scf	73	218
Product Gas, vol%	Adjusted for 450,000 Bur/hr Heat Loss and N ₂ Purge Free	
H ₂	13.7	31.0
CO	7.7	18.2
CH ₄	2.3	7.8
CO ₂	14.3	35.0
N ₂	62.0	8.0
Total	100	100
Heating Value, Btu/scf	96	239
Sulfur Retention %	80.4	72.7

Table 26. TRDU Operating Conditions and Gasification Efficiency Results for P073

Test	- 10 mesh Blacksville	- 10 mesh Blacksville
Oxidant	Air	Oxygen
Gasifier Temp., °C	950	922
Coal/Sorbent Feed Rate, lb/hr	273	285
Air Flow, lb/hr	924	145
O ₂ Flow, lb/hr	0	168
Steam Flow, lb/hr	167	290
Air:Coal Ratio, lb/lb	4.23	0.63
Steam:Coal Ratio, lb/lb	0.76	1.27
O ₂ :Coal Ratio, lb/lb	0.98	0.87
Recirculation Rate, lb/hr	13185	11265
Operation Pressure, psig	120	120
TRDU Riser Velocity, ft/s	48.2	34.3
Carbon Conversion		
Solid Accountability	74.5	73.4
Noncondensable Gas Make	57.3	64.6
Thermal Oxidizer	82.2	87.8

Table 27. XRFA of the Pittsburgh No. 8 Bituminous Coal Ashes

Element, wt%	P073	P073	P073	P073	P073	P073
	SP 12:20 04/23/03	FV 12:25 04/23/03	SP 07:00 04/24/03	FV 7:15 04/24/03	SP 19:18 04/29/03	FV 19:25 04/29/03
Si	86.2	36.2	51.7	22.3	87.2	38.9
Al	0.9	5.8	3.3	8.0	1.0	7.4
Fe	1.7	5.3	5.1	8.5	1.7	6.7
Ti	0.1	0.3	0.2	0.4	0.1	0.3
P	0.0	0.1	0.1	0.1	0.0	0.1
Ca	5.8	30.3	19.2	35.8	4.9	25.8
Mg	3.6	14.4	10.9	16.5	3.2	12.3
Na	0.0	0.1	0.0	0.1	0.0	0.2
K	0.3	0.6	0.5	0.7	0.5	0.9
S	1.2	6.9	8.9	7.6	1.5	7.4
Total	99.8	100	99.9	100	100.1	100

Table 28. Corrected TRDU Product Gas Compositions for TRDU Test P074

Test	Falkirk	Falkirk	Falkirk–Wood	Falkirk–Wood
Product Gas Composition, vol%	Air	Oxygen	Air	Oxygen
H ₂	8.3	18.8	7.2	15.7
CO	6.5	6.9	5.7	8.6
CH ₄	1.1	2.7	1.2	2.5
CO ₂	12.9	28.8	12.4	24.9
N ₂	74.7	42.6	70.9	47.3
Total	103.4	99.9	97.4	99.0
Heating Value, Btu/scf	59	111	54	104
% N ₂ in Feed	29.8	62.1	32.8	69.6
N ₂ -Free Heating Value, Btu/scf	75	166	81	194
Product Gas, vol%	Adjusted for 450,000 Btu/hr Heat Loss and N ₂ Purge Free			
H ₂	16.9	34.8	18.5	34.1
CO	13.1	12.7	14.6	18.7
CH ₄	2.2	5.1	3.1	5.4
CO ₂	14.7	36.7	18.2	37.6
N ₂	53.0	10.7	45.6	4.1
Total	100	100	100	100
Heating Value, Btu/scf	120	205	138	226

Table 29. TRDU Operating Conditions and Gasification Efficiency for TRDU Test P074

Test	Falkirk	Falkirk	Falkirk/Wood	Falkirk/Wood
Oxidant	Air	Oxygen	Oxygen	Oxygen
Gasifier Temp., °C	822	798	863	839
Coal/Sorbent Feed Rate, lb/hr	453	502	381	463
Air Flow, lb/hr	860	0	990	0
O ₂ Flow, lb/hr	0	213	0	212
Steam Flow, lb/hr	95	416	121	341
Steam:Coal Ratio, lb/lb	0.21	0.92	0.35	0.82
O ₂ :MAF Coal Ratio, lb/lb	0.84	0.87	1.17	1.07
Recirculation Rate, lb/hr	2255	4005	9045	8550
TRDU Riser Velocity, ft/s	54.8	42.4	43.5	39.8
Carbon Conversion				
Solid Accountability	90.0	85.3	96.5	93.2

Table 30. XRFA of Falkirk Lignite and Falkirk–Wood Samples

Element, wt%	Falkirk	Falkirk	Falkirk–Wood	Falkirk–Wood
	SP	FV	SP	FV
Si	22.9	29.9	69.5	35.7
Al	6.1	11.5	8	5.6
Fe	4.9	9	4.2	5.3
Ti	0.4	0.6	0.3	0.3
P	0.1	0.1	0	0.1
Ca	56.5	35.8	9.6	34
Mg	3.9	6.2	4.3	16.1
Na	2.2	3.6	2.5	1.6
K	1.6	2	1.6	1.4
S	1.4	1.3	0	0
Total	100	100	100	100.1

low melting species such as potassium were not building up in the bed material. Steady-state data from these tests are also given in Appendix A.

4.4.9 TRDU Gasification Test P075

TRDU Test P075 tested the thermally dried Australian brown coals from the Loy Yang and Lochiel Mines over the week of December 1, 2003, through December 4, 2003. This test generated 59 hours of coal feed and 46 hours of gasification, including 33 hours of air-blown gasification and 13 hours oxygen-blown gasification. Tables 31 and 32 summarize some results from the brown coal tests and compare them to previous tests conducted with the high-sodium North Dakota lignite from the Freedom Mine. Table 33 shows the ash chemistry from selected samples from the gasification

Table 31. Corrected TRDU Product Gas Compositions for TRDU Test P075 Utilizing Australian Brown Coal as Compared to North Dakota Lignite

Coal	Freedom	Freedom	Loy Yang	Loy Yang	Lochiel	Lochiel
Product Gas Composition, vol%	Air	Oxygen	Air	Oxygen	Air	Oxygen
H ₂	7.7	17.3	7.4	9.3	6.3	13.8
CO	4.5	5.7	5	6.0	4.6	4.9
CH ₄	1.1	2.2	1.7	3.1	1.5	3.5
CO ₂	12.6	30.0	12.7	22.0	13.9	24.7
N ₂	73.4	47.6	73.5	51.6	74.9	56.0
Total	99.3	102.8	100.3	98.0	101.2	102.9
Heating Value, Btu/scf	50	96	57	81	50	96
% N ₂ in Dry Feed	35.3	62.7	32.0	74.0	32.8	74.8
N ₂ -Free Heating Value, Btu/scf	71	141	84	195	70	192
Product Gas, vol%	Adjusted for 450,000 Btu/hr Heat Loss and N ₂ Purge Free					
H ₂	19.1	33.9	17.9	25.8	13.1	33.2
CO	11.0	11.2	12.1	16.6	9.6	11.8
CH ₄	2.6	4.2	4.1	8.6	3.1	8.4
CO ₂	16.2	40.0	18.5	40.7	19.5	39.1
N ₂	51.1	10.9	47.5	8.3	54.8	7.4
Total	100	100	100	100	100	100
Heating Value, Btu/scf	124	188	138	224	105	231

Table 32. TRDU Operating Conditions and Gasification Efficiency Results for P075 Utilizing Australian Brown Coal as Compared to North Dakota Lignite

Coal	Freedom m	Freedom	Loy Yang	Loy Yang	Lochiel	Lochiel
Oxidant	Air	Oxygen	Air	Oxygen	Air	Oxygen
Gasifier Temp., °C	815	782	882	795	785	741
Coal/Sorbent Feed Rate, lb/hr	354	469	267	443	479	482
Air Flow, lb/hr	758	0	766	0	1004	0
O ₂ Flow, lb/hr	0	201	0	192	0	168
Steam Flow, lb/hr	80	489	125	319	124	293
Steam:Coal Ratio, lb/lb	0.24	1.3	0.46	0.72	0.31	0.61
O ₂ :maf Coal Ratio, lb/lb	0.96	0.97	0.78	0.51	0.86	0.50
Recirculation Rate, lb/hr	2905	1225	11,880	2215	8305	11,225
TRDU Riser Velocity, ft/s	33.5	50.8	35.8	35.8	41.8	30.3
Carbon Conversion Solid Accountability	84.3	80.9	77.3	83.0	72.0	74.4

tests with the high-sodium Freedom lignite and the high-sodium Australian brown coals. All-steady state data are given in Appendix A.

4.4.10 TRDU Gasification Test P076

TRDU Test P076 tested as-received Wyodak subbituminous coals from the Belle Ayr Mine over the period of June 1, 2004, through June 8, 2004. This test generated 180 hours of coal feed and 176 hours of gasification, including 101 hours of air-blown gasification and 75 hours oxygen-blown gasification. Tables 34 and 35 summarize some results from the Wyodak tests. Wyodak was selected for testing based on its excellent performance in the previous TRDU tests and in testing at the PSDF. These tests were conducted primarily to look at establishing a baseline for mercury emissions and measuring the performance of various Hg sorbents.

4.4.11 TRDU Gasification Test P077

TRDU Test P077 tested raw and washed subbituminous coals from India over the week of September 21, 2004, through September 25, 2004. This test generated 82 hours of coal feed and 72 hours of gasification, including 60 hours of air-blown gasification and 13 hours oxygen-blown gasification. Table 36 summarizes the range of operating conditions and results achieved with both Wyodak subbituminous coal and the high-ash Indian subbituminous coal tests conducted recently. Wyodak was selected as the comparison coal since it is the baseline coal for both the EERC and PSDF transport reactors and it is the closest in rank to the Indian coal tested.

Tables 37 and 38 summarize the identified steady-state periods which had the best performance for both the recent Wyodak and high-ash Indian coal tests. Typically, the best performance is a balance of fuel gas heating value and carbon conversion; however, with the low fuel gas heating values being achieved with the high-ash Indian coals, more of an emphasis was placed on fuel gas heating value at the expense of carbon conversion.

Table 33. XRFA of Selected Samples from Gasification of High-Sodium Low-Rank Coals

Element, wt%	Freedom SP	Freedom FV	Loy Yang SP	Loy Yang FV	Lochiel SP	Lochiel FV
Si	16.4	10.3	88.6	47.5	79.2	10.7
Al	4.9	6.4	4.3	5.4	4.2	6.2
Fe	5.1	4.6	2.1	6.4	2.3	6.4
Ti	0.3	0.4	0.1	0.3	0.2	0.7
P	0.1	0.2	0	0.1	0	0
Ca	42.7	47.1	2.4	13.3	7.2	30.6
Mg	25.0	26.5	0.8	12.5	2.3	25.3
Na	3.2	2.8	1.5	6.6	3.9	6.7
K	0.9	0.5	0.2	1.9	0.1	0.6
S	1.4	1.2	0	5.9	0	12.7
Total	100	100	100	99.9	99.4	99.9

Table 34. Corrected TRDU Product Gas Compositions for TRDU Test P076 Utilizing Wyodak PRB Subbituminous Coal

Coal	Wyodak	Wyodak	Wyodak	Wyodak	Wyodak	Wyodak
Product Gas Composition, vol%	Air	Oxygen	Air	Oxygen	Air	Oxygen
H ₂	6.6	15.8	7.5	15.5	8.1	15.6
CO	4.4	8.5	5.1	7.0	6.2	7.2
CH ₄	1.1	2.9	1.5	2.9	1.6	2.8
CO ₂	11.6	21.9	12.2	21.0	13.6	22.0
N ₂	70.7	49.3	71.2	49.9	74.1	49.9
Total	94.4	98.4	97.5	96.3	103.6	97.8
Heating Value, Btu/scf	47	108	56	102	62	102
% N ₂ in Dry Feed	33.9	71.4	34.1	71.6	31.9	72.5
N ₂ -Free Heating Value, Btu/scf	78	200	87	200	90	197
Product Gas, vol%	Adjusted for 450,000 Btu/hr Heat Loss					
H ₂	17.8	34.0	17.6	35.5	17.7	34.6
CO	11.9	18.3	12.0	16.0	13.6	16.0
CH ₄	3.0	6.2	3.5	6.6	3.5	6.2
CO ₂	19.4	31.0	19.0	31.0	20.0	33.5
N ₂	48.0	10.4	47.9	11.0	45.3	9.7
Total	100	100	100	100	100	100
Heating Value, Btu/scf	126	233	131	234	136	226

Tables 39 and 40 show ash analyses from samples taken at the end of each operating condition (air versus oxygen). These analyses show that the circulating standpipe material was still enriched in silica from the start-up sand. However, the filter ash seems to be very representative of the coal ash plus the small amount of dolomite being fed into the TRDU. This is very consistent with previous testing in which the filter ash chemistry was shown to be very representative of the coal/dolomite ash chemistry from within a few hours of starting up the gasifier. It typically takes a hundred hours or more to flush most of the silica sand from the circulating bed material.

Oxygen-blown operation requires the addition of considerable excess steam to maintain the reactor temperatures below the temperature where ash deposition and agglomeration of the circulating ash material become a problem. Figure 8 is a plot of both the corrected dry and wet product gas heating values and carbon conversion for the two subbituminous coals tested under both air- and oxygen-blown conditions. Carbon conversion seems to be primarily dependent on the ratio of weight oxygen fed to the weight of the maf coal fed regardless of what form the oxygen was fed (air versus oxygen). The oxygen-to-maf coal ratio was considerably higher for the high-ash Indian coal because of the higher gasifier operating temperature and the extra heat needed to bring all of the coal ash up to bed temperature. The corrected dry product gas heating value for the oxygen-blown test has a significantly higher HHV than air-blown operation (190–225 Btu/scf as compared to 90–120 Btu/scf). A comparison of the fuel gas heating values shows that the Indian coal had corrected heating values that were as high as 90 Btu/scf. This heating value would be marginal for

Table 35. TRDU Operating Conditions and Gasification Efficiency Results for P076 Utilizing Wyodak PRB Subbituminous Coal

Coal	Wyodak	Wyodak	Wyodak	Wyodak	Wyodak	Wyodak
Oxidant	Air	Oxygen	Air	Oxygen	Air	Oxygen
Gasifier Temp., °C	890	886	844	848	868	869
Coal/Sorbent Feed Rate, lb/hr	285	334	325	325	363	337
Air Flow, lb/hr	906	0	933	0	986	0
O ₂ Flow, lb/hr	0	241	0	216	0	213
Steam Flow, lb/hr	121	315	124	302	120	298
Steam:Coal Ratio, lb/lb	0.44	0.98	0.40	0.97	0.34	0.92
O ₂ :maf Coal Ratio, lb/lb	1.094	1.070	0.988	0.985	0.935	0.938
Recirculation Rate, lb/hr	3585	11,815	7280	11,780	6595	5640
TRDU Riser Velocity, ft/s	41.0	40.0	42	38	41	38
Carbon Conversion						
Solid Accountability	95.0	97	92	95	93	97

Table 36. TRDU Range of Operating Conditions for Air-Blown Operation

Coal Name Coal Type	Wyodak Subbituminous	As-Mined Indian Subbituminous	Washed Indian Subbituminous
Moisture Content, %	23	4.5	9
Pressure, psig	120	120	120
Steam–maf Coal Ratio	0.49–0.72	0.68–0.77	0.62–0.82
O ₂ –maf Coal Ratio	0.8–1.1	1.0–1.2	1.0–1.2
Ca–S Ratio, mole (sorbent only)	1.5	1.4	1.4
Coal and Sorbent Feed Rate, lb/hr	285–415	416–465	382–505
Avg. Mixing Zone Temp, °C , avg.	844–894	879–948	928–936
HHV of Fuel Gas, act., Btu/scf	47–66	41–43	33–50
HHV of Fuel Gas, cor., Btu/scf	126–140	80–89	84–90
Conversion, %	87–94.5	80–89	84–90
Carbon in Bed, %, Standpipe	1.6–5.0	1.1–2.5	1.7–2.5
Riser Velocity, ft/s	38.1–43.8	49.7–51.1	48.2–55.6
Standpipe Velocity, ft/s	0.20–0.25	0.20–0.22	0.21–0.22
Circulation Rate, lb/hr	3590–8840	2610–7520	3340–4350
Total Operating Hours	175.5		65*

* Total for as-mined and washed Indian subbituminous.

operating a gas turbine without either enriched air operation or some supplemental fuel source. These lower heating values and carbon conversions as compared to the low-ash Wyodak coal are thought to be due to a lower char/steam reactivity and to the extra heat loss associated with heating all of the ash up to bed temperature. Typically, the lower heating value of the fuel gas stream achieved from air-blown and oxygen-blown gasifiers would have approximately the same heating value entering the gas turbine combustor since the high volume of steam addition needed in the oxygen-blown system acts like the nitrogen diluent that would enter the gas turbine combustor under air-blown operation. This high steam addition to the oxygen-blown transport reactor is necessary to prevent the formation of hot zones in the circulating bed material where bed material agglomeration and deposition can occur. Generally, the similar fuel gas heating values entering a gas turbine make it hard to justify the economics of an oxygen-blown transport reactor strictly for power production. However, in this case, some oxygen enrichment or oxygen operation may be necessary in order to achieve fuel gas heating values sufficient to guarantee operation of the gas turbine. In addition, concepts such as a Vision 21 plant in which a gasifier would be operated for both power and fuels or chemicals production could justify the higher capital and operating costs associated with an oxygen plant.

4.4.12 Conclusions of Gasification Testing

In total, eleven test campaigns utilizing the L-valve loop seal configuration have been conducted under enriched air- or full oxygen-blown conditions. During these tests, 1515 hours of coal feed with 660 hours of air-blown gasification and 720 hours of enriched air- or oxygen-blown gasification were completed. During these tests, approximately 366 hours of operation with Wyodak,

Table 37. TRDU Operating Conditions for Best-Case Gasification Performance

Test	Wyodak	Wyodak	Raw Indian	Raw Indian	Washed Indian	Washed Indian
Oxidant	Air	Oxygen	Air	Enriched Air	Air	Oxygen
Gasifier Temp, °C	848	848	936	946	936	895
Coal/Sorbent Feed Rate, lb/hr	359	325	448	451	441	461
Air Flow, lb/hr	968	0	1160	689	1087	0
O ₂ Flow, lb/hr	0	216	0	101	0	249
Steam Flow, lb/hr	129	303	172	215	168	332
Steam:Coal Ratio, lb/lb	0.37	0.97	0.40	0.50	0.40	0.75
O ₂ /maf Coal Ratio, lb/lb	0.93	0.99	1.10	1.06	1.07	1.01
Recirculation Rate, lb/hr	8840	11,780	3280	6460	3780	4630
TRDU Riser Velocity, ft/s	43.5	37.5	50.6	46.4	48.2	41.1
Carbon Conversion	92.2	94.6	87.7	86.8	87.7	84.5

Table 38. Actual and Corrected TRDU Product Gas Compositions for Best-Case Steady-State Tests

Test	Wyodak	Wyodak	Raw Indian	Raw Indian	Washed Indian	Washed Indian
Oxidant	Air	Oxygen	Air	Enriched Air	Air	Oxygen
Product Gas Composition, vol%						
H ₂	8.2	15.5	5.6	8.6	6.6	13.4
CO	5.4	7.0	4.1	5.6	4.5	7.3
CH ₄	1.5	2.9	1.2	1.7	1.4	2.8
CO ₂	11.7	21.0	12.8	17.8	13.6	22.6
N ₂	71.2	49.9	73.9	65.0	76.2	55.9
Total	98.6	96.3	97.6	98.7	102.3	102.0
Heating Value, Btu/scf	59	102	44	63	50	95
% N ₂ in Dry Feed	33.2	71.6	25.9	33.0	26.8	70.0
N ₂ -Free Heating Value, Btu/scf	91	200	57	87	62	179
Product Gas, vol%			Adjusted for 450,000 Btu/hr Heat Loss			
H ₂	18.6	35.5	10.5	16.7	11.7	29.2
CO	12.3	16.0	7.7	10.9	8.0	15.9
CH ₄	3.4	6.6	2.2	3.3	2.5	6.1
CO ₂	17.4	31.0	15.8	23.8	16.0	32.9
N ₂	48.3	11.0	63.8	45.3	61.8	15.9
Total	100	100	100	100	100	100
Heating Value, Btu/scf	135	234	81	123	90	208

Table 39. XRF Analysis of TRDU Samples Generated from the Raw High-Ash Indian Coal, %

	Raw Indian Coal Ash	Raw Indian – Air		Raw Indian – Enriched Air	
		Standpipe	Filter	Standpipe	Filter
Si	56.7	79.3	51.6	81.6	51.7
Al	29.1	12.6	27.5	10.6	27.8
Fe	6.2	3.4	6	3.2	6
Ti	2.8	0.6	2.2	0.5	2.3
P	0.5	0.1	0.5	0.1	0.5
Ca	1.7	1.8	6.6	2	6.3
Mg	0.7	0.8	3.5	0.8	3.1
Na	0.6	0.7	0.6	0.6	0.6
K	1.7	0.7	1.6	0.5	1.6
S	0	0	0	0	0
Total	100	100	100.1	99.9	99.9

Table 40. XRF Analysis of TRDU Samples Generated from Washed High-Ash Indian Coal, %

	Washed Indian Coal Ash	Washed Indian – Air		Washed Indian – Oxygen	
		Standpipe	Filter	Standpipe	Filter
Si	55.4	83.2	48.3	80.4	50.4
Al	29.6	8.9	25.7	11.6	27.1
Fe	7	3.2	5.6	3	5.4
Ti	2.5	0.4	2.1	0.6	2.3
P	0.1	0.1	0.6	0.2	0.6
Ca	2.1	2.1	10.1	2.1	9
Mg	0.7	1	5.5	1	3.1
Na	0.7	0.6	0.5	0.6	0.5
K	1.8	0.4	1.5	0.6	1.6
S	0	0	0	0	0
Total	99.9	99.9	99.9	100.1	100

123 hours of operation with Navajo subbituminous coal, 143 hours of operation on Illinois No. 6, 106 hours on SUFCo, 110 hours on Prater Creek, 48 hours on Calumet, and 134 hours on a Pittsburgh No. 8 bituminous coal were completed. In addition, 331 hours of operation on low-rank coals such as North Dakota lignite, Australian brown coal, and a 90:10 wt% mixture of lignite and wood waste were completed. Also included in these test campaigns was 50 hours of gasification on a petroleum coke from the Hunt Oil Refinery. An additional 73 hours of gasification on a high-ash Indian coal was generated.

Figures 8 and 9 show the product gas heating value and the carbon conversion measured as a function of the O_2/maf coal ratio for the various ranks of coal under both air-blown and oxygen-blown testing, respectively. In general, operation on the more reactive low-rank western coals has displayed higher carbon conversions and product gas heating values even when operating at lower reactor temperatures than comparable bituminous coal tests. From Figures 8 and 9 it is apparent that the transport gasifier performs better on the lower-rank feedstocks because of their higher char reactivity with the gasification reactions. Comparable carbon conversions have been achieved at similar oxygen/coal ratios for both air-blown and oxygen-blown operation. Figure 10 shows the operating results from the TRDU on the low-rank coals only as a function of the O_2/maf coal ratio. This figure shows that these low-rank feedstocks provided similar fuel gas heating values; however, even among the low-rank coals, there was still some variability in the carbon conversion. The carbon conversion was lower for the fuels (brown coal in particular) that contained a significant amount of fines. Appendix A contains all of the steady-state operating conditions and process performance data calculated for each steady-state period identified.

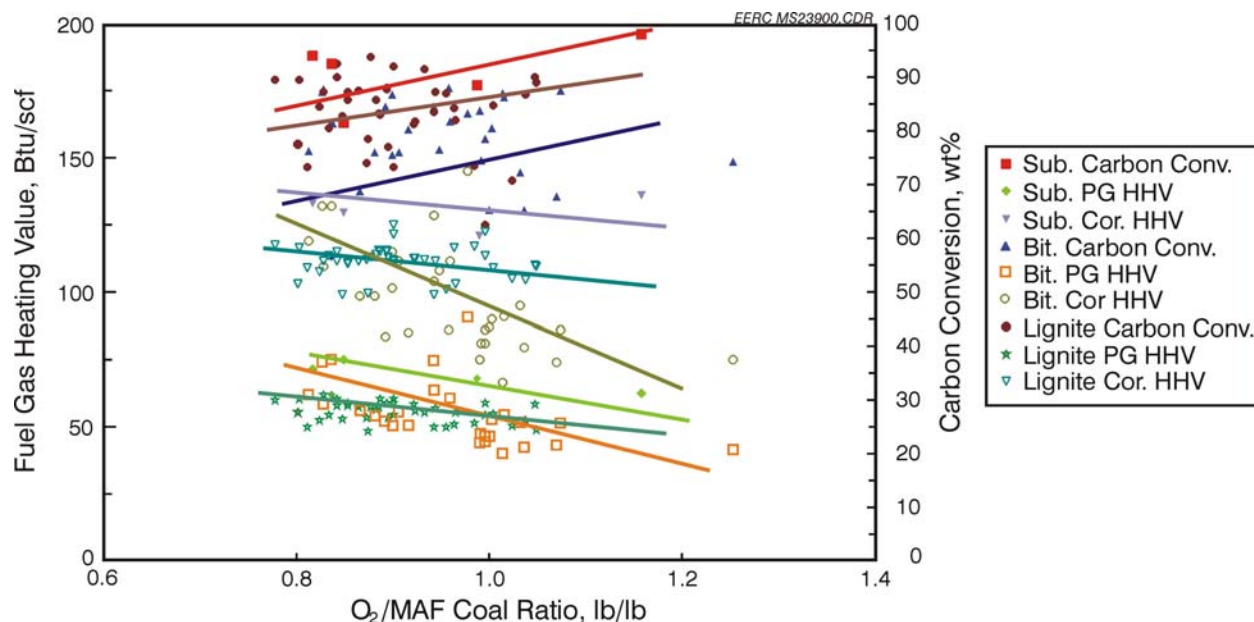


Figure 8. Effect of coal rank as a function of O_2/maf coal ratio on carbon conversion and fuel gas heating value under air-blown operation in the TRDU.

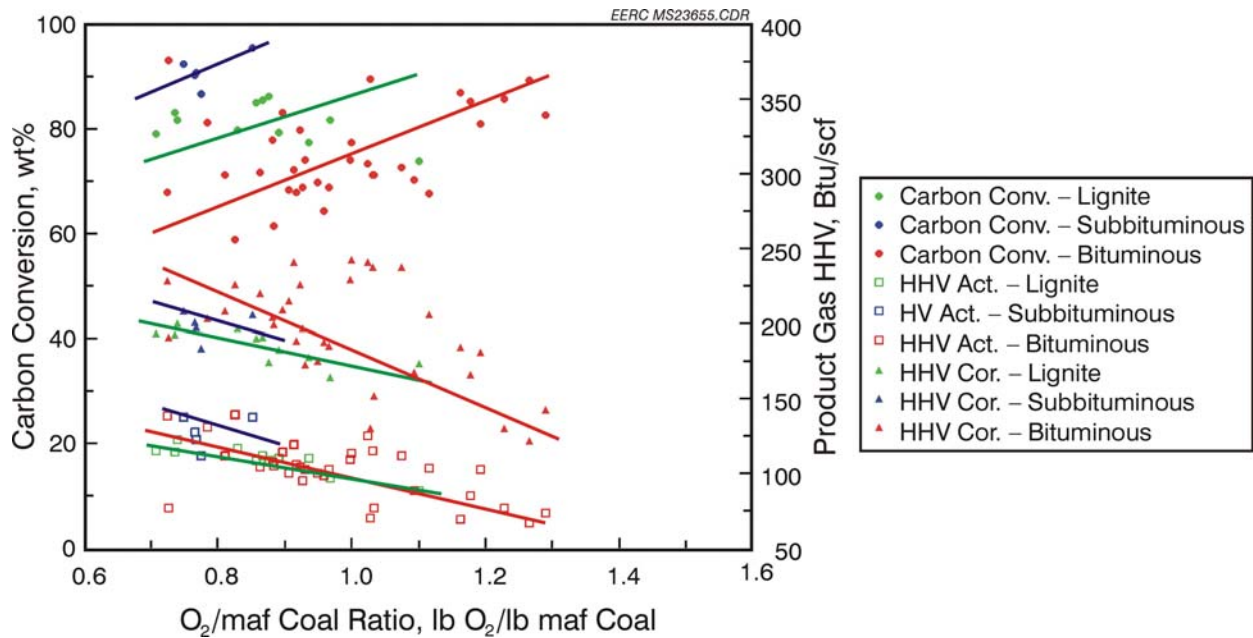


Figure 9. Effect of coal rank as a function of O₂/maf coal ratio under oxygen-blown operation in the TRDU.

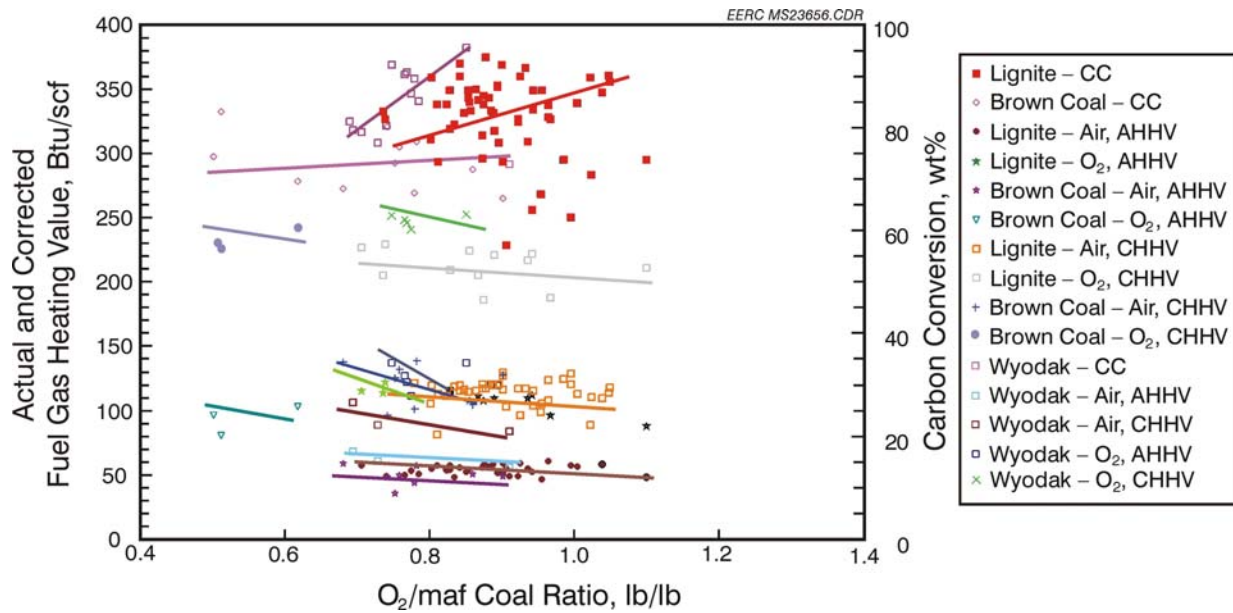


Figure 10. Effect of O₂/maf coal ratio on low-rank coal performance in a transport gasifier (CC = carbon conversion, AHHV = actual higher heating value, CHHV = corrected higher heating value).

The more reactive lower-rank fuels had higher carbon conversions and corrected dry product gas heating values than the higher-rank bituminous coals. The bituminous coals were operated at higher oxygen/maf coal ratios than the lower-rank coals since they typically were operated at higher reactor temperatures in an effort to achieve the same level of steam gasification. For all fuels, carbon conversion increased and corrected dry product gas heating value decreased with an increasing oxygen/maf coal ratio. Appendix A contains all of the steady-state operating conditions and process performance data calculated for each steady-state period identified.

4.5 Hot-Gas Filter Vessel Operation

No major failures of these candles have occurred in over 2500 hours of testing with approximately 2175 hours in gasification mode. The HGFV has mostly been operated between 460° and 570°C (860° and 1058°F) at a face velocity of approximately 3.8 to 4.5 ft/min. Backpulse operating parameters were 270 to 400 psig reservoir pressure with either 1/4- or 1/2-second pulse valve opening times. The average particulate loading going into the HGFV has ranged from approximately 4500 up to 45,000 ppm with a d_{50} between 7 to 22 μm , depending on the fuel type, quantity of sorbent utilized for sulfur control, and whether solids were being recirculated from the dipleg back into the standpipe. A substantial increase in the “cleaned” filter baseline (from ~40 to >90 inches H₂O) has been observed in a few of the tests. This filter ash has averaged from 25 to 60 wt% carbon depending on the carbon conversion and has a low bulk density of approximately 20 lb/ft³. The small size, the lack of the cohesiveness seen in other filter ashes, and the low density of the ash suggests that a high percentage of the filter cake will be reentrained back onto the filters after they are backpulsed. More details about the hot-gas filter performance have been given elsewhere (3, 7).

In gasification mode, the pulse frequency has been short, with pulses occurring every 8 to 15 minutes. This rapid pulsing is thought to be due to the high-carbon, low-density dust with a high aerodynamic drag being able to minimize the porosity of the filter cake on the surface of the candle. This results in a rapid rise in pressure drop across the filters. The data acquisition system on the TRDU has been programmed to save the filter vessel differential pressure and the filter outlet static pressure every 2 seconds whenever a backpulse sequence is started until 30 seconds after the last manifold is backpulsed.

Operation of the HGFV during the last two gasification tests P076 and P077 utilized ten 1.5-m Pall Advanced Separation iron aluminide candle filters exclusively. The HGFV was operated between 190° and 300°C (375° and 575°F) at a face velocity of approximately 2.2–3.2 ft/min. Backpulse operating parameters were approximately 360 psig backpulse reservoir pressure with a 0.5-s opening time. The average particulate loading going into the HGFV ranged from approximately 4500 up to 38,000 ppm, with a d_{50} between 9 and 12 μm with a top size 95% less than 40 μm (see Figure 11). Figure 11 also shows the particle-size distribution for these same samples for both the circulating bed material (standpipe) and filter ash along with the particle-size distribution of the feed coal. This figure shows that the coal had an average feed size of approximately 400–500 μm with less than 15 wt% being less than 100 μm . The circulating bed material was approximately 200 μm in size as compared to the 160- μm average size of the silica sand. The filter ash particle-size distribution (~10 μm for the raw Indian coal and slightly larger ~12 μm for the washed Indian coal)

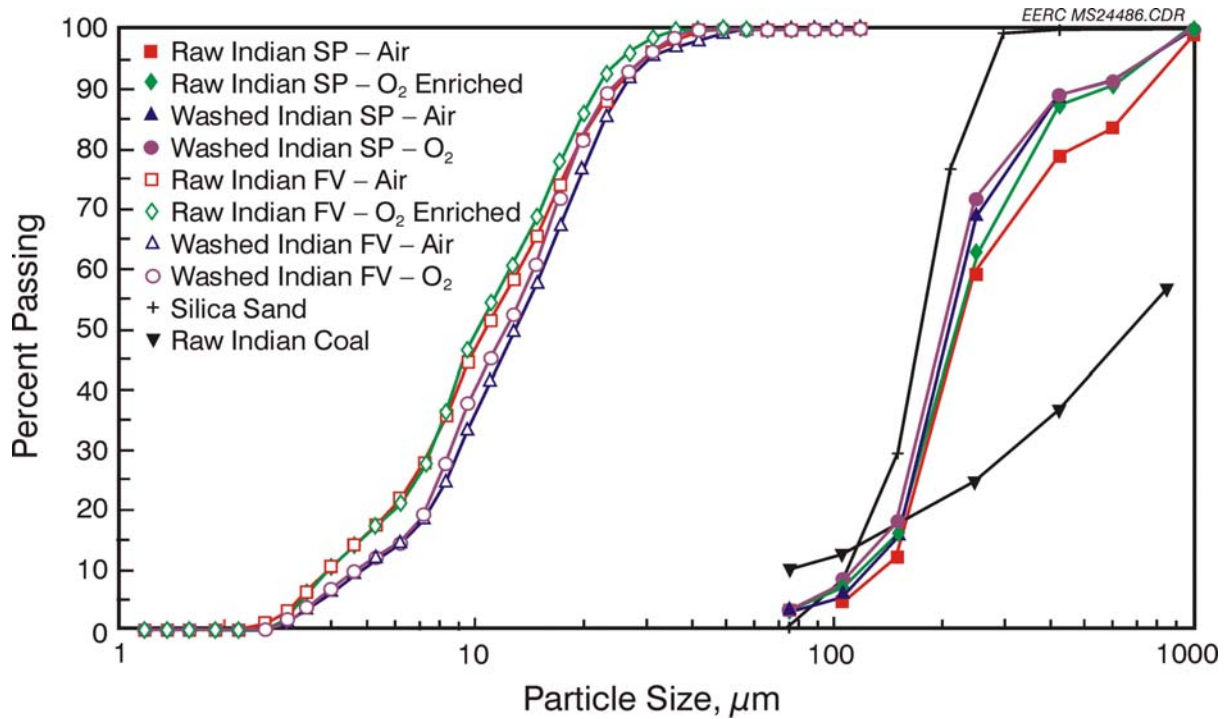


Figure 11. Particle-size distribution of high-ash Indian coal samples from a transport reactor.

were both below the filter particle-size distribution typically achieved with Wyodak subbituminous coal. The smaller particle-size distribution possibly could be due to the higher loadings of more dense ash particles into the cyclones, resulting in a better cyclone efficiency even for the less dense carbon particles in the circulating bed material. Outlet dust loadings were maintained at 1 ppmw or below, indicating good performance from the iron aluminide candle filters.

4.6 Measurement of Mercury in TRDU Fuel Gas

One goal of the transport reactor project has been to demonstrate the acceptable performance of mercury continuous emission monitors (CEMs) to measure mercury in actual coal-derived fuel gas. The EERC is attempting to evaluate the form of mercury (ionic, elemental, or particulate-bound) on the TRDU gasifier. Testing has involved both wet-chemistry methods and mercury CEMs. For the mercury CEMs, three different pretreatment systems could be used to determine which system gives the best results. The first uses a basic stannous chloride solution to convert all mercury to the elemental form and remove any gases such as HCl that may result in interferences. The second CEM uses an acid stannous chloride solution for conversion but has a heated alkali trap to remove interfering gases. The third system that could be used is a thermal system with dilution. Sampling on the TRDU facility has been done at one location on the outlet of the hot-gas filter system. The pressure at this point is approximately 116 psig. Wet-chemistry sampling has consisted of a modified EPA Method 29 multimetal trains to look for mercury in each sample. A detailed description of this method can be found at EPA's Web page: www.epa.gov/ttn/emc. In addition, three different types of mercury CEMs have been used to measure mercury continuously at this sampling location. The wet-chemistry samples were used to

verify that the mercury CEM is giving good results. The EERC performs the wet-chemistry mercury analyses on-site so that the results can be obtained quickly for comparison purposes and quality control/quality assurance(QA/QC).

4.6.1 Description of Mercury CEM

4.6.1.1 Semtech Hg 2010 Instrument

The commercial Semtech Hg 2010 mercury analyzer is essentially a portable Zeeman-modulated cold-vapor atomic absorption spectrometer (CVAAS) that can monitor Hg^0 continuously. By using an online reduction unit, total mercury can be monitored continuously. In the reduction unit, a reducing solution (SnCl_2) is pumped to the sampling probe. The extracted gas sample and reducing solution are transported continuously through a mixing spiral to maximize the gas solution residence time and ensure complete conversion of Hg^{2+} to Hg^0 . After conversion of all the mercury to Hg^0 , the sample gas is transferred to a Peltier cooled gas-liquid separator. The conditioned dry gas is then analyzed by the instrument using CVAAS techniques. To minimize interferences from the presence of H_2S , hydrocarbons, and fine particulate in the flue gas sample, the analyzer uses Zeeman effect background correction by applying a modulated magnetic field to a mercury lamp.

4.6.1.2 OhioLumex RA-915+

The OhioLumex RA-915+ is a real-time continuous monitor for total and elemental mercury measurement. The instrument is based on differential Zeeman atomic absorption spectroscopy using high-frequency modulation of light polarization. A mercury lamp is placed in a permanent magnetic field which has the ability to slightly change the wavelength of the mercury light. This allows for background correction for such broadband absorbers as SO_2 , moisture, and particulate matter. The Lumex has a multipass cell which provides an effective path length of 10 meters. The instrument does not use gold amalgamation preconcentration, which allows for a faster response time. In ambient air, a lower detection limit of 2 ng/m^3 can be achieved according to the manufacturer. OhioLumex provides a cell for thermal reduction of oxidized mercury to elemental mercury. No catalyst is used in the thermal decomposition cell.

The Lumex needs an external mercury supply such as a permeation device or a gas cylinder for calibration. The instrument does come with a small cell of fixed volume that contains saturated mercury vapor which can be used to check the calibration.

An earlier version of the OhioLumex instrument was evaluated during the first round of the U.S. Environmental Protection Agency (EPA) Environmental Technology Verification Program (8).

4.6.1.3 P S Analytical Sir Galahad and Tekran Model 2537

The P S Analytical and Tekran mercury CEMs are very similar in operation. Both instruments use a batch process where mercury is collected on a specialized gold trap and then desorbed into an atomic fluorescence (AF) analyzer. The primary difference between the two is the type of gold trap that is used. In both cases, the exact manufacturing technique is proprietary. The P S Analytical

instrument was initially developed and used for the natural gas industry and the Tekran for ambient mercury measurements. For both instruments, between 0.5 and 2 L/min of flue gas (depending on mercury concentration) is pumped through a gold trap, which is maintained at a constant temperature. Once the mercury has been adsorbed on the gold trap, the trap is removed from the flue gas stream and flushed with argon. The mercury is then desorbed from the gold trap at 500°C using a heating coil. The mercury is then carried to the AF analyzer using argon as a carrier gas. Once the mercury has been desorbed from the trap, it is rapidly cooled with additional argon. To speed up the measurement process, a dual gold trap is used. As one trap is adsorbing mercury, the second trap is being desorbed. The approximate time for each measurement is 2–5 minutes. The operating mercury concentration range for AF-type mercury CEMs is up to 5 orders of magnitude. They can measure mercury concentrations from about 1 ng/m³ to 150 µg/m³, making these instruments ideal for measuring the low concentrations (<5 µg/m³) often found in flue gas generated from coal-fired systems.

Both the Tekran and P S Analytical CEMs are calibrated using Hg⁰ as the primary standard. The Hg⁰ is contained in a closed vessel which is held in a thermostatic bath. The temperature of the mercury is monitored, and the amount of mercury is calculated using vapor pressure calculations. Currently, the calibration of the Tekran is more automated; however, the calibration of the P S Analytical CEM is also being automated. Typically, the calibration of the units has proven stable over a 24-hour period. The EERC has spent considerable effort to develop a sample conditioning procedure that provides representative results.

4.6.2 Hg Sampling Results

Since all of the mercury has been shown to be in the elemental form, it was hoped that very little sample conditioning would be required. However, sampling after the sample gas conditioning system utilized for the other gas analyzer indicated that no mercury was reaching the mercury CEM. Tests with water-filled impingers also exhibited issues with obtaining representative samples. Sample conditioning with basic stannous chloride solutions appeared to work over the short term; however, the reducing fuel gas would consume the reagents in the solution and affect the CEM readings. The sample conditioner was then set up with a peristaltic pump to pump fresh solution into the impingers and pump spent solution out of the impingers. This sample conditioning system has worked well in providing a fuel gas that continuously has worked well with the Hg CEM; however, this sample conditioning requires frequent human intervention to add fresh solution and remove the spent solution. It also generates a fair amount of waste material.

After resolving the sample conditioning issues, shakedown testing with four different types of analyzers (P S Analytical Sir Galahad, Semtech Hg 2010, Nippon DM6B, and the OhioLumex RA-915+), were tested. The light signal from the AA- based analyzers Semtech, Nippon, and OhioLumex appeared to be attenuated when the fuel gas was run through them. The Semtech and Nippon mercury analyzers were unsuccessful because some component of the fuel gas attenuated the signal to such a degree that no measurement could be made, while the OhioLumex appeared to trend the mercury emissions; however, a fourfold change in the factory calibration factor was required to the data to accurately trend the measured emissions. Further use of this analyzer may be warranted after consulting with OhioLumex to understand a scientific reason for the change in the calibration factor. After considerable testing and the addition the P S Analytical analyzer

modification adopted during testing at the TECO coal gasification plant, the EERC has had good success measuring the mercury concentrations in the warm fuel gas. Based on these results, the EERC selected the P S Analytical Sir Galahad for further testing on the TRDU. This is the instrument that uses two gold traps in series to trap mercury. The first trap is exposed to the conditioned fuel gas sample at room temperature. After the sample is collected, the first trap is heated with an air purge to drive off the mercury and combust any collected organic material. This gas stream is passed through the second gold trap which captures the mercury. The second trap is then desorbed with an argon carrier gas, and the mercury is measured in the atomic fluorescence chamber.

The initial sample conditioner configuration consisted of an insulated sample line passing through the gasification tower wall to a area that was not explosion-rated. After passing through the wall, the flow was split. One line carried excess flow to a vent and was controlled with a needle valve. This line was used to maintain a flow rate high enough to ensure no condensation in the sample line. The second line was connected to a Teflon filter holder and a heated Teflon sample line. The filter was maintained at temperature with a heated muff set to 163°C (325°F). The flow through this line is also controlled with a needle valve. The sample line was connected to a set of stainless steel impingers. The first impinger was filled with a 20% NaOH/2% SnCl₂ solution to reduce all forms of mercury to the elemental state. This impinger was maintained at room temperature to prevent the solution from freezing. The second impinger was dry and maintained in an ice bath to remove moisture and other condensibles. The third and fourth impingers were dry and maintained in a cold bath of isopropanol and dry ice to ensure all heavier tars and oils were condensed. The last three impingers were filled with glass marbles to enhance heat transfer. A P S Analytical gold trap at room temperature was connected to the outlet of the last impinger. After the gold trap, the flow passed through a dry gas meter to measure the total dry sample volume.

The first several attempts to load the gold traps resulted in loadings that were too high for the P S Analytical Sir Galahad to measure. An acceptable sample volume of 0.10 ft³ was obtained by continuing to half the total sample volume until the concentration was within the calibrated range of the PSA. The results are summarized in Table 41.

It appears there were problems with the first sample. The remaining samples are fairly consistent until the sample conditioning chemicals were consumed.

For the testing in April 2003, the sample conditioning equipment was modified to use a needle valve for pressure letdown before the Teflon-coated stainless steel impingers. The fuel gas and SnCl₂/NaOH solution were continuously filled with peristaltic pumps and mixed in a Teflon "T" just prior to entering the first impinger. The first impinger was left in water (not cooled). The second impinger was also in water and was only chilled by conduction from the next chamber in the impinger box. The remaining three impingers were chilled to roughly -6.7°C (20°F), -5°C (20°F [12°-23°F]) in a glycol bath. Liquids were removed from the second and third impingers with a second peristaltic pump. Most of the moisture was taken out in the second impinger.

Table 41. Initial Hg CEM Data Collected Utilizing the P S Analytical Sir Galahad Hg CEM

Sample	Date	Time	Sample Volume, ft ³	Hg Measured, pg	Hg Concentration, µg/m ³
1	10/23/02	5:45 pm	0.1	80,792.282	28.5
2	10/23/02	6:00 pm	0.1	41,906.1074	14.8
3	10/23/02	6:28 pm	0.05	17,585.792	12.4
4	10/23/02	6:57 pm	0.101	43,241.3843	15.1
5	10/23/02	7:12 pm	0.099	57,117.7956	20.4
6	10/23/02	7:12 pm	0.104	41,160.6491	14
7	10/23/02	7:12 pm	0.102	50,082.8335	17.3
8	10/23/02	7:26 pm	0.1	49,661.6426	17.5

For the test in September, a purged cabinet (for maintaining an explosion-proof rating with non-explosion-proof equipment) was installed on the seventh deck of the TRDU tower to hold the two peristaltic pumps, the supply and waste jugs for the SnCl₂ solution, and the chiller for the impingers. The configuration of the impingers was:

1. Gas and SnCl₂/NaOH solution in water at room temperature used to reduce oxidized mercury to elemental mercury.
2. Gas-liquid separator in water at room temperature used to remove condensibles.
3. Gas-liquid separator chilled to remove condensibles.
4. and 5. Chilled impingers to remove condensibles.

An unheated 1/4-inch Teflon line was used to transport the sample gas to the P S Analytical Sir Galahad mercury analyzer in the third-floor lab.

For the December 2003 sampling, the major changes for the sample conditioning system were to plumb a vent line for excess flow in after the pressure letdown valve but before the sample line leading to the Hg CEM. This was done to increase the gas flow through the pressure letdown valve to keep it warm and prevent plugging but to avoid bringing a large volume of gas that may contain traces of organics through the sample conditioner and reduce the chance that some might break through to the Hg CEM. In addition, a second 1/4-inch Teflon line was plumbed from the third-floor lab up to the impingers to allow spiking of elemental mercury into the impingers from the P S Analytical CAVKIT. The line was plumbed in upstream of the SnCl₂/NaOH line. A check valve and a shut-off valve were also plumbed in to prevent pressurized flow back to the CAVKIT. Five Teflon-lined stainless steel impingers were used. All were filled approximately 3/4 full with glass marbles to enhance heat transfer. The first impinger was used as a gas-liquid contactor. It was placed in cool water to help quench the fuel gas. The second impinger was used as a gas-liquid separator. It was placed in a water bath, which was somewhat cooled by contact with a chilled water bath. The fourth

impinger was also used as a gas–liquid separator to remove any condensate. The third, fourth, and fifth impingers were in a chilled water bath that was maintained between -8° and -1°C (17° and 30°F).

On December 1, 2003, sampling of TRDU fuel gas began at 11:40 and continued until 16:30. The concentration varied throughout the day, ranging from 2.5 to $10\ \mu\text{g}/\text{m}^3$ (P S Analytical) as shown in Figure 12. On December 5, 2003, Figure 13 shows the Hg CEM and Method 29 data while both the Sir Galahad and the Lumex analyzers were used for sampling. The TRDU was switching between air-blown and oxygen-blown modes during the late morning. Two Method 29 samples were completed before 11:00. The measured concentrations from the CEMs are significantly higher than the Method 29 values. All CEM data were corrected for CO_2 removal.

Table 42 shows the wet-chemistry methods utilized to determine the total and speciated Hg found in the fuel. Except for the first test which was probably sampled for too long, thereby consuming all of the permanganate solution, the amount of mercury recovered in the KCl and 20% H_2O_2 solutions (i.e., oxidized forms of mercury) was very small. Essentially 90% or better of the mercury is in the elemental form in the fuel gas.

Some of the best data obtained on the monitoring and removal of mercury in the fuel gas was generated during Test P076 utilizing the baseline Wyodak subbituminous coal.

After the sample conditioning issues were resolved, with shakedown testing on three different types of analyzers (P S Analytical Sir Galahad, Semtech Hg 2010, and the OhioLumex RA-915+), the EERC selected the P S Analytical Sir Galahad for further testing on the TRDU. The light signal from the Semtech and OhioLumex both appeared to be attenuated when the fuel gas was run through them. The Semtech attenuated off-scale while the OhioLumex appeared to trend the mercury emissions; however, a fourfold change in the factory calibration factor was required to accurately trend the measured emissions. Further use of this analyzer may be warranted after consulting with OhioLumex to understand the change in the calibration factor. After considerable testing and adding the P S Analytical analyzer modification adopted during testing at the TECO coal gasification plant, the EERC has had good success measuring the mercury concentrations in the warm fuel gas. Figure 14 shows the Hg CEM measurements obtained with the P S Analytical Sir Galahad against those obtained utilizing the modified EPA Method 29 wet chemistry. This graph shows good agreement between the two methods. Figure 15 shows the results when the EERC-treated carbon/limestone mixture was injected in to the HGFV over a 2.5-hour period (10:00–12:30). The treated carbon was mixed with the limestone to make it more flowable and to allow the feeder to be operated at higher motor speeds. The baseline mercury CEM measurement dropped from approximately $26\ \mu\text{g}/\text{m}^3$ to approximately $18\ \mu\text{g}/\text{m}^3$ or a 30% reduction in the mercury (Hg^0) emissions. In-duct injection tests at higher temperatures around 350°C (662°F) appeared to have only a small effect on mercury emissions. This treated carbon feed rate was selected to give comparable feed rates for the treated carbon as utilized in other EERC-conducted pilot- and field-scale testing on combustion systems. These combustion tests showed that over 90% capture of Hg^0 was possible at similar sorbent feed rates on a volumetric basis as what was tested in the TRDU HGFV (5). This reduced performance of the treated activated carbon is probably affected by several variables, including the higher ash loading/carbon loading to the HGFV than combustion baghouses or electrostatic precipitators

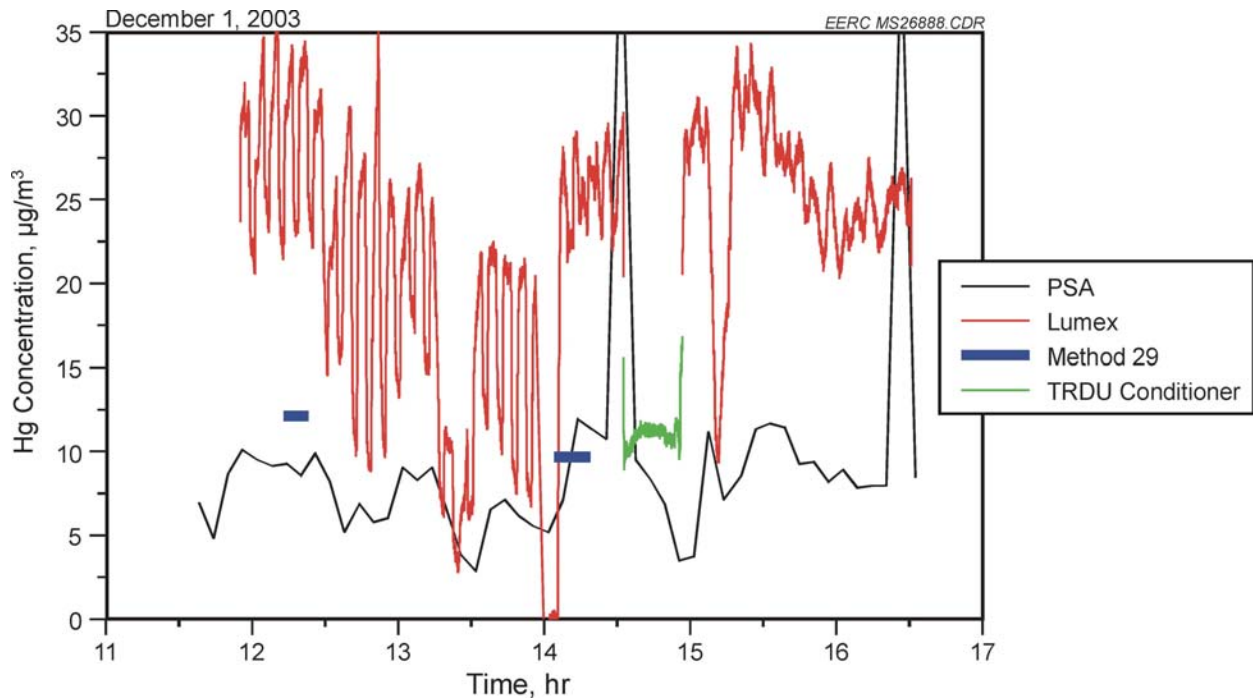


Figure 12. Comparison of Hg CEM and Method 29 data for Australian Lochiel brown coal.

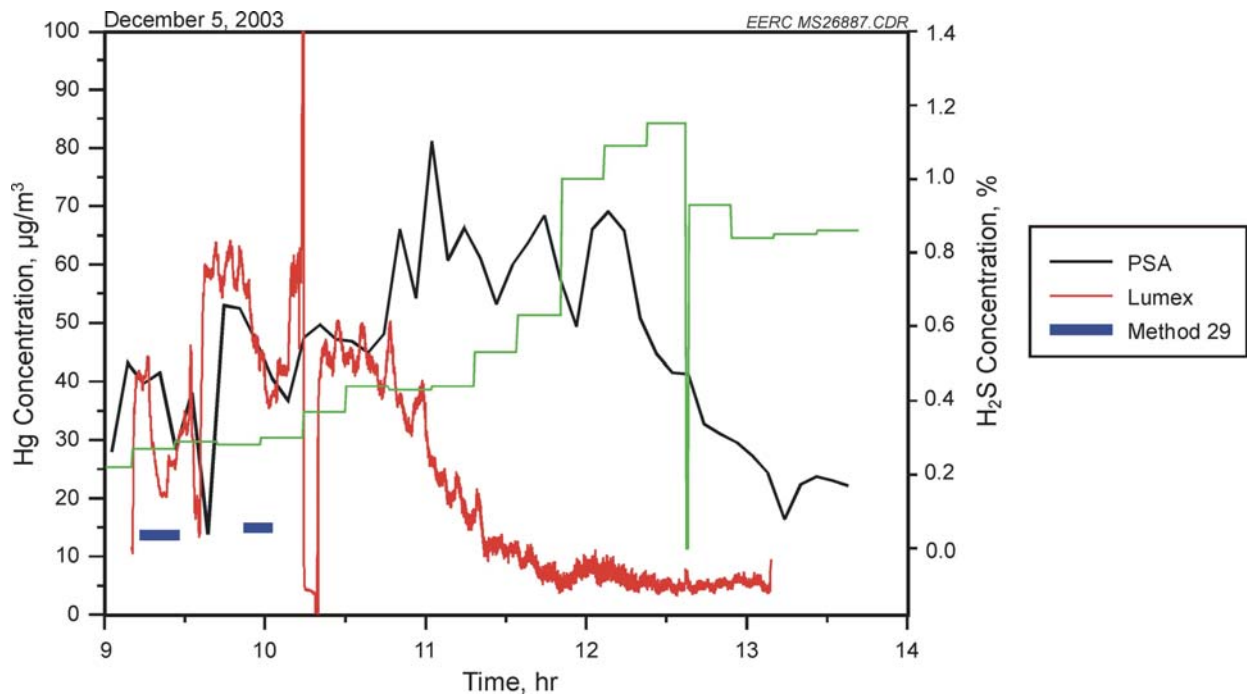


Figure 13. SEM micrograph of mixing zone deposit from gasification Test P060 on SUFCo coal showing Points 1–4.

Table 42. Results from Ontario Hydro/EPA Method 29 Data

Date	KCl Conc., $\mu\text{g}/\text{m}^3$	20% H_2O_2 Conc., $\mu\text{g}/\text{m}^3$	KMnO_4 Conc., $\mu\text{g}/\text{m}^3$	Total Conc., $\mu\text{g}/\text{m}^3$
37738	1.14	0.02	0.02	1.18
37738	0.4	0.26	3.17	3.84
37891	1.36	0	9.61	10.97
37955	NM*	0.7	11.42	12.14
37955	NM	0.32	9.35	9.67
37959	NM	0	13.85	13.7
37959	NM	0	14.86	14.86

* Not measured as part of Method 29 train, Ontario Hydro train only.

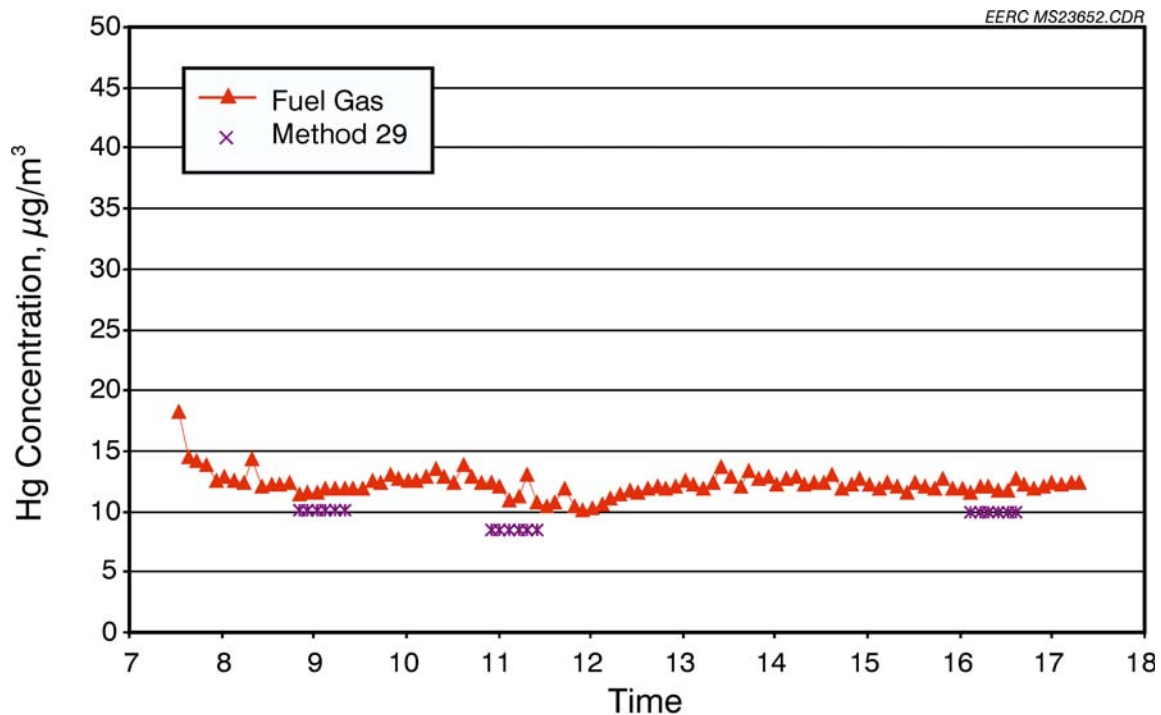


Figure 14. Comparison of Hg CEM with Method 29 wet-chemistry data on TRDU fuel gas.

(ESPs); the presence of different impurities such as H_2S , tars, and possibly NH_3 ; and the higher operating temperatures of the filter system as compared to the combustion systems.

Similar EERC-conducted combustion tests where the additive was added directly to the coal and nontreated carbon was fed into the baghouse or ESP had also exhibited mercury control greater than 90%. This type of test was also attempted with the TRDU under gasification conditions.

Figure 16 shows the results from this test where the active part of the additive was added to the coal starting at 9:30 in the morning; however, the addition of the additive alone did not appear to significantly change the mercury concentration in the fuel gas. At 4:12 in the afternoon, activated carbon feed was started from the feeder to the HGFV. This carbon was fed over a 3-hour and 40-minute period until a problem with the peristaltic pump delivering the basic stannous chloride solution to the sample conditioning traps failed, thereby terminating the test. The data shown in Figure 16 indicate that the presence of the additive together with the activated carbon injection was removing the mercury from approximately $24 \mu\text{g}/\text{m}^3$ down to approximately $15 \mu\text{g}/\text{m}^3$ (37.5% removal), and the trend still seemed to be dropping when the test was terminated.

A packed-bed system utilizing a slipstream of warm TRDU fuel gas was also constructed and tested. This packed-bed system was designed to run a slipstream of approximately 1000 scfh of fuel gas through a 3-inch-diameter and 15-inch-deep bed of sorbent. Three different tests utilizing a coarse EERC-treated activated carbon (F2BO and F2HO) were conducted. One additional test utilizing an amended silicate sorbent from ADA Technologies was also tested in the packed-bed contactors. Figures 17–19 show the breakthrough curves for these sorbents being tested at approximately 265°C (510°F). The one test with the EERC-treated carbon at a higher temperature exhibited very little mercury removal. Figures 17 and 18 show the breakthrough curve for the EERC-treated carbon with both a slow and fast heatup rate, respectively. There did not appear to be any major difference in the sorbent performance since both sorbents had breakthrough times of approximately 1.5 hours; however, the sorbent with the fast heatup time seemed to remove the mercury from the starting baseline of $26 \mu\text{g}/\text{m}^3$ to less than $1 \mu\text{g}/\text{m}^3$ while the slow-heatup sorbent only reduced mercury to approximately $2.5 \mu\text{g}/\text{m}^3$. Tests with a smaller quantity of the ADA Technologies sorbent UP-EB-X015 mixed with silica sand were conducted to measure mercury breakthrough. The silica sand selected had a very high pressure drop (> 100 psid), resulting in a maximum fuel gas flow rate through the bed of 315 scfh. This test shown in Figure 19 indicates that the mercury level was dropped from approximately $28 \mu\text{g}/\text{m}^3$ to approximately $3.5 \mu\text{g}/\text{m}^3$ for approximately 0.5 hour.

The EERC has spent considerable effort to develop a sample-conditioning procedure that provides representative results. Since all of the mercury has been shown to be in the elemental form, it was hoped that very little sample conditioning would be required. However, sampling after the gas-conditioning system utilized for the other gas analyzer indicated that no mercury was reaching the mercury CEM. Tests with water-filled impingers also exhibited issues with obtaining representative samples. Sample conditioning with basic stannous chloride solutions appeared to work over the short term; however, the reducing fuel gas would consume the reagents in the solution and affect the CEM readings. The sample conditioner was then set up with a peristaltic pump to pump fresh solution into the impingers and pump spent solution out of the impingers. This sample-conditioning system has worked well in providing a fuel gas that works well with the Hg CEM; however, this sample conditioning requires frequent human intervention to add fresh solution and remove the spent solution. It also generates a fair amount of waste material that needs to be dealt with.

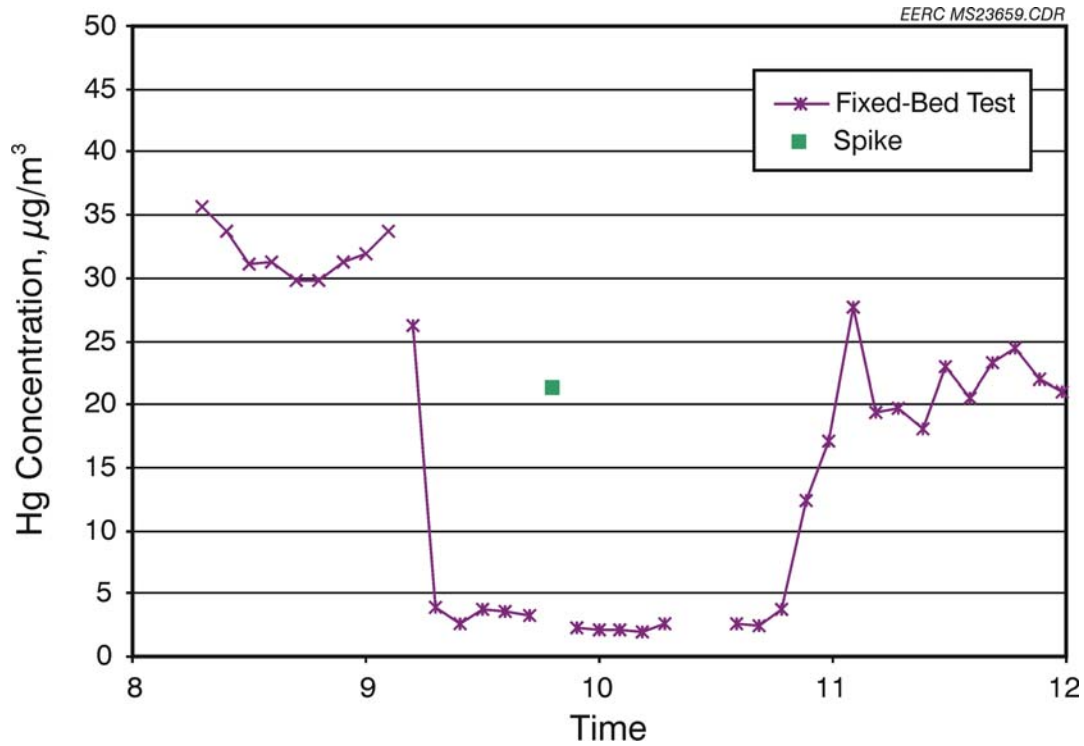


Figure 15. Effect of EERC-treated carbon injection on Hg removal in hot-gas filter vessel.

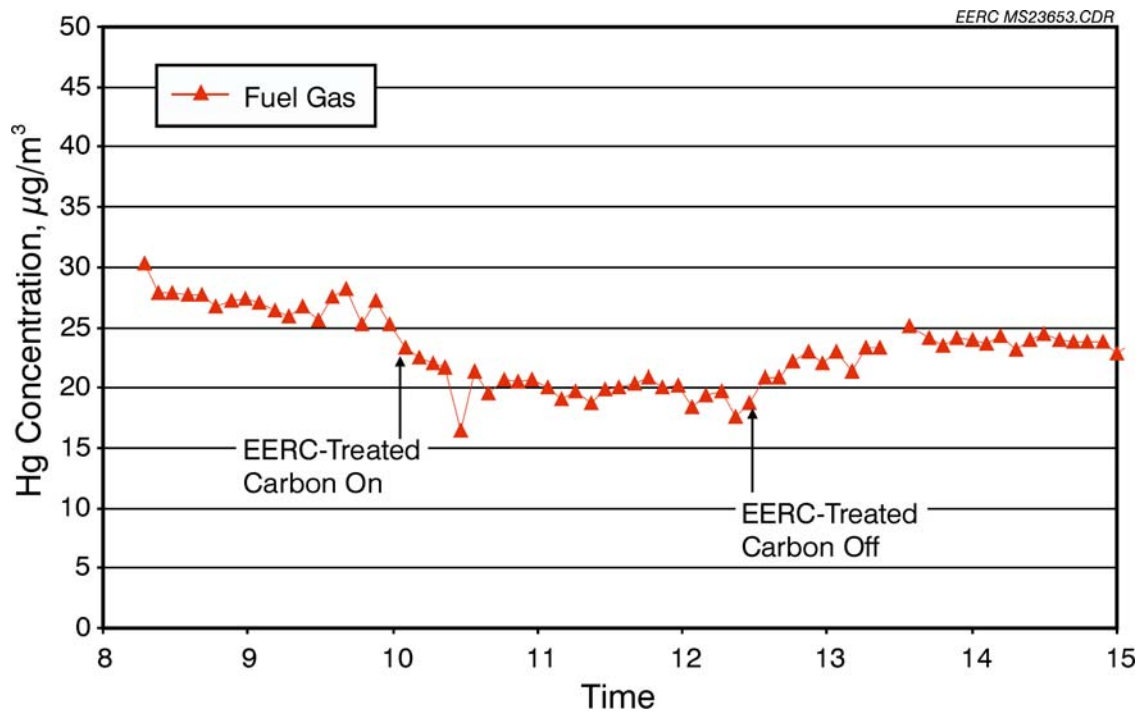


Figure 16. Effect of additive addition to Wyodak coal and activated carbon injection to hot-gas filter vessel on Hg removal.

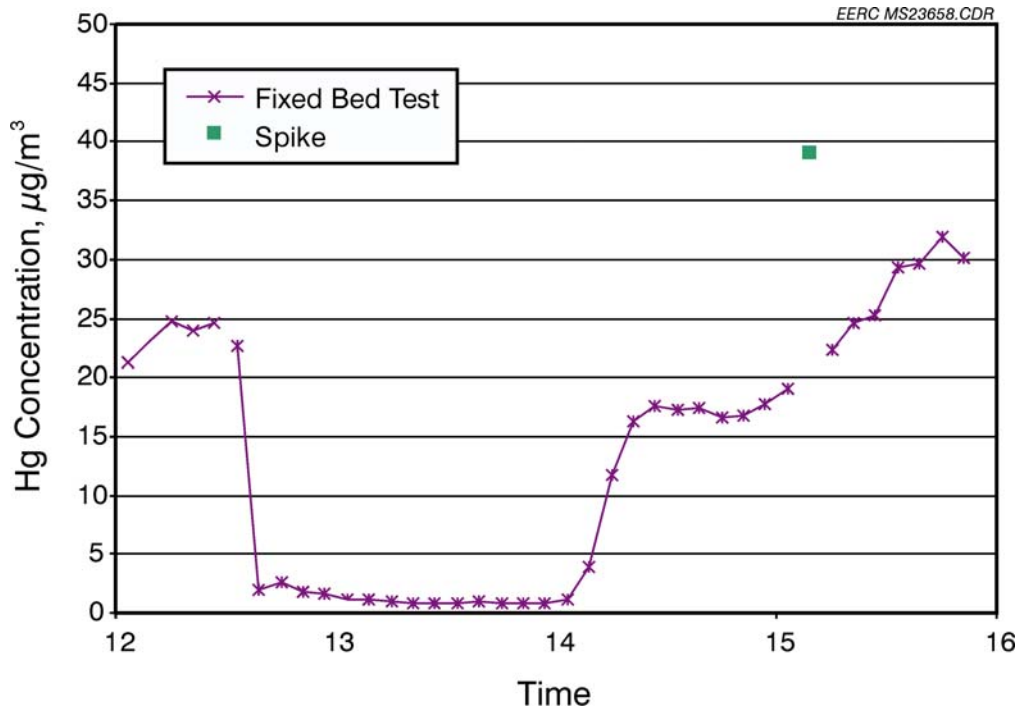


Figure 17. Packed-bed Hg sorbent removal with EERC-treated carbon at 265°C (510°F).

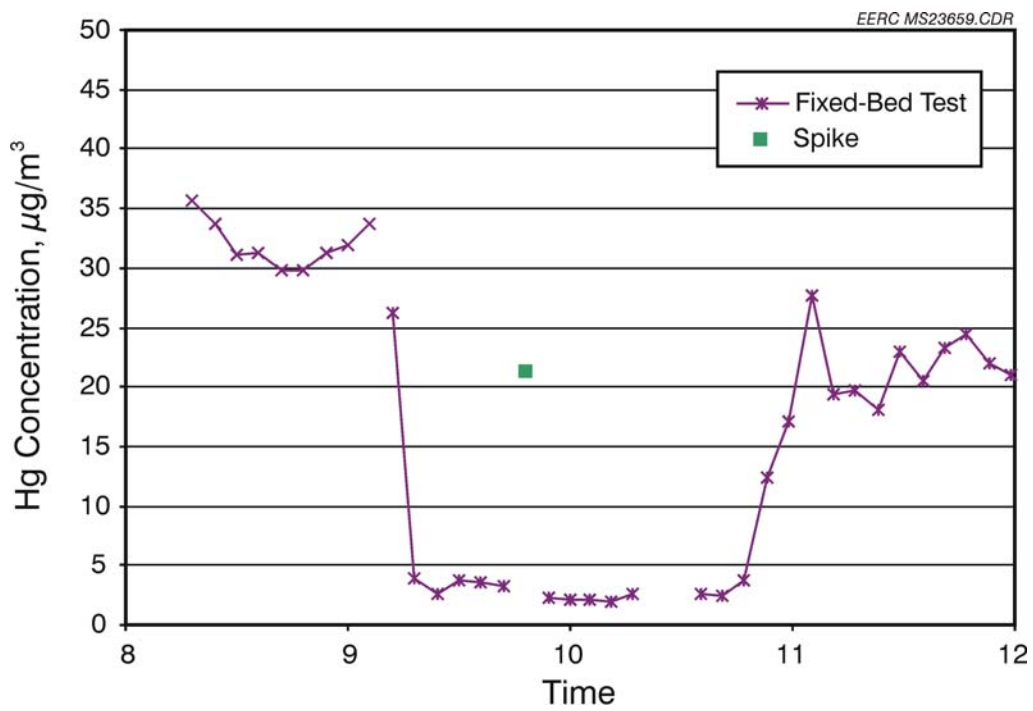


Figure 18. Hg removal with packed-bed test utilizing EERC-treated carbon at 265°C (510°F).

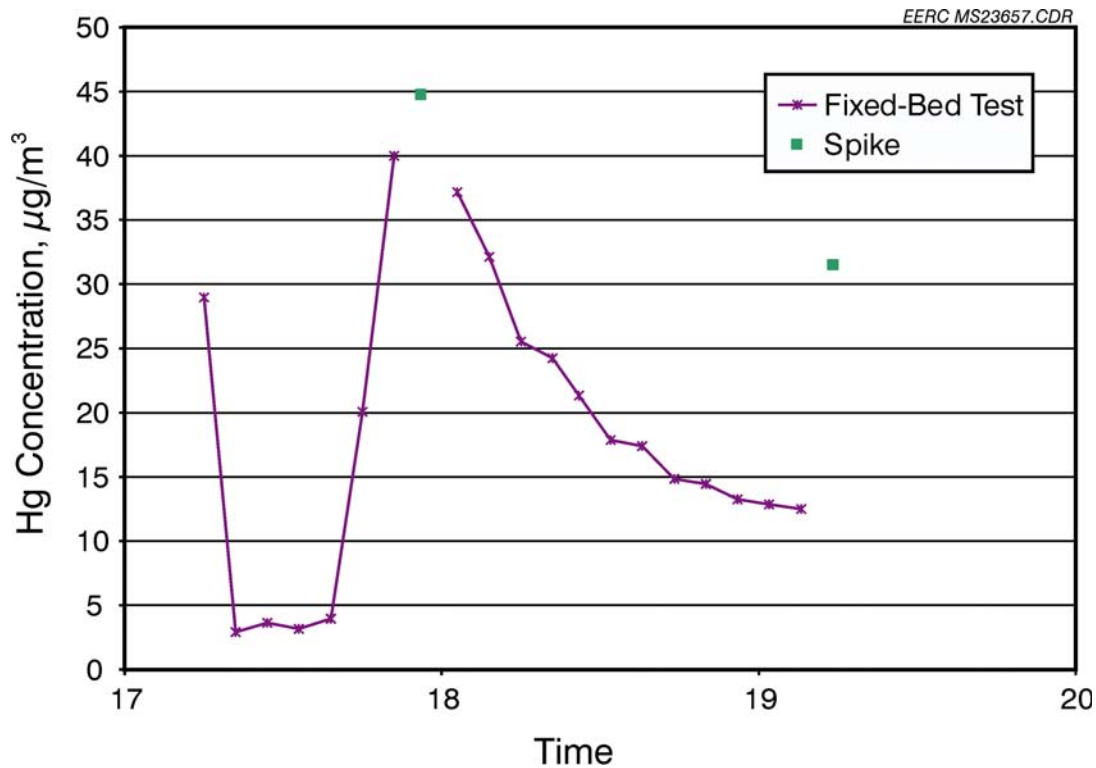


Figure 19. Hg removal test with packed-bed reactor utilizing ADA Technologies sorbent UP-EB-X015 at 260°C (500°F).

Figures 20–22 show the Hg CEM measurements with the P S Analytical Sir Galahad against those obtained utilizing the modified EPA Method 29 wet chemistry or at the outlet of the packed-bed slipstream test stand for the 3 days of operation on the Indian coal. These graphs shows that the mercury emissions for the Indian coal seemed to have lined out after a conditioning period at 15–20 µg/Nm³ in air-blown mode and increased up to 40–50 µg/Nm³ in oxygen-blown mode. The wet chemistry also showed that this mercury was almost exclusively elemental in nature with no oxide forms of mercury being detected.

A packed-bed system utilizing a slipstream of warm TRDU fuel gas was also tested. Figure 22 shows the results obtained while utilizing a coarse EERC-treated activated (F2ZO) carbon. This particular sorbent reduced the mercury emissions to less than 5 µg/Nm³ for up to 4 hr at 215°C. The overall mercury removal was approximately 90% with a space velocity of 3060 hr⁻¹. This sorbent still had not exhibited a definitive breakthrough at the end of this test.

4.7 TRDU Deposit Formation

The TRDU was modified after Test P059 to increase the mixing zone diameter in an effort to operate at lower mixing zone velocities. Lower velocities result in longer residence times for the solid carbon and improved gasification kinetics. Lower velocities could also result in increased

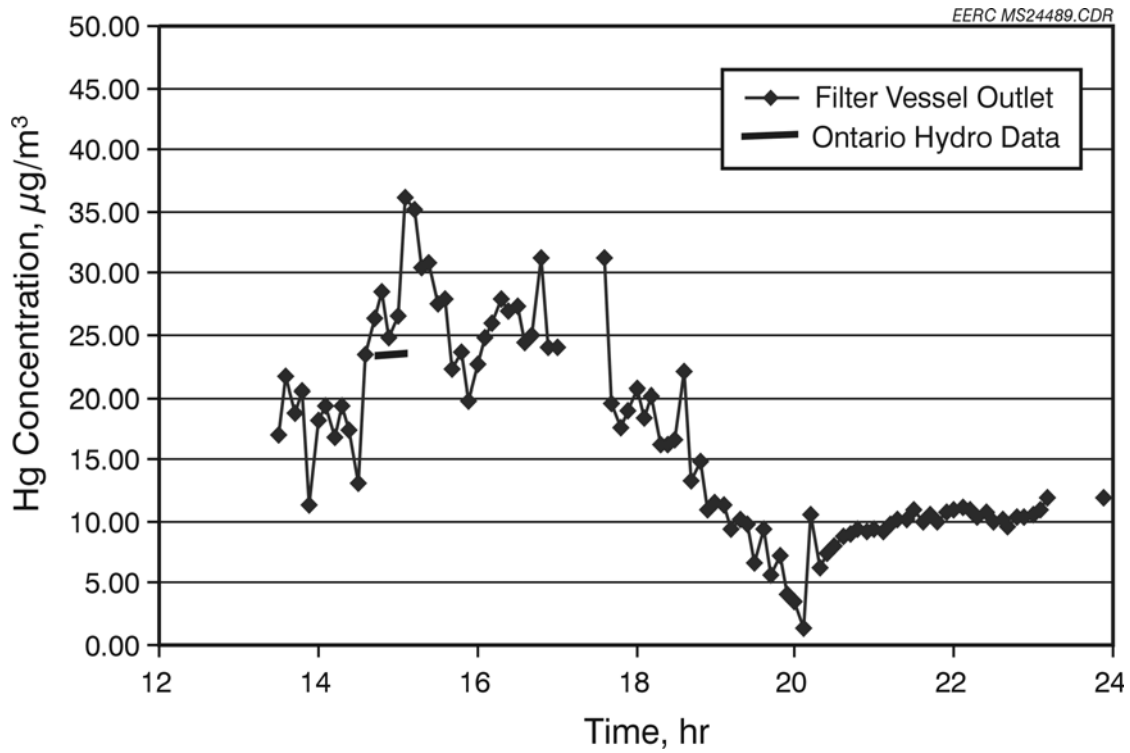


Figure 20. Hg CEM measurements, October 19.

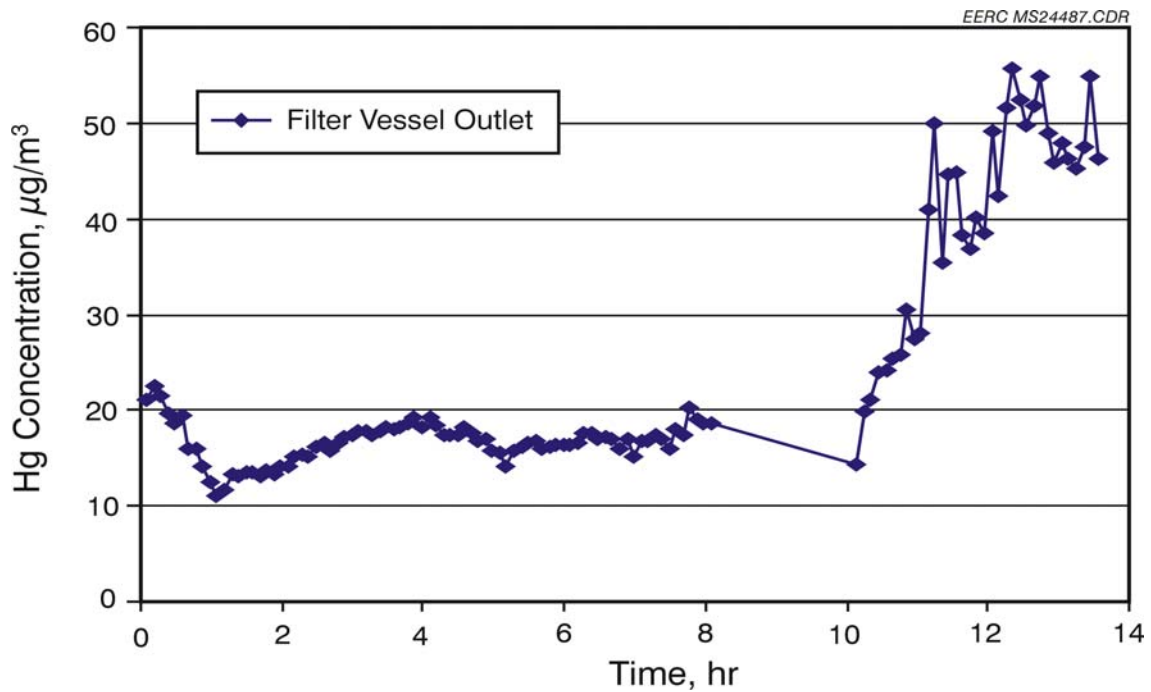


Figure 21. Hg CEM measurements, October 20.

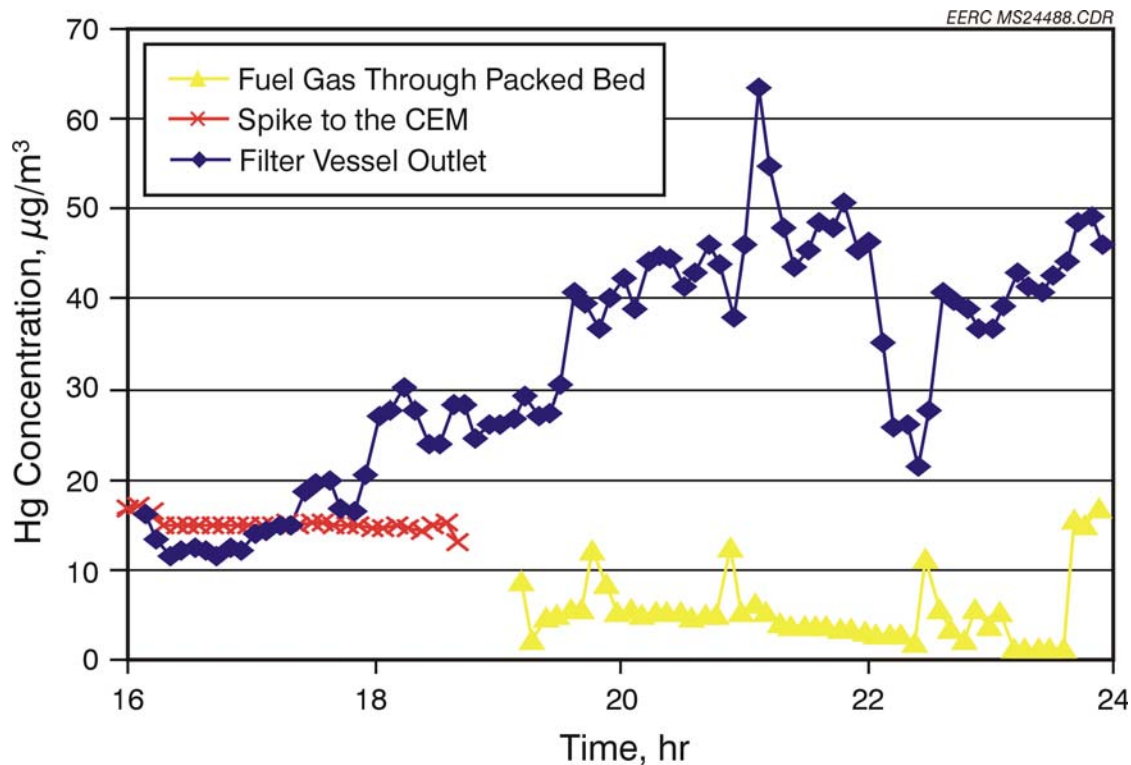


Figure 22. Hg CEM measurements, October 21.

occurrences of bed material agglomeration and deposition because of the presence of more localized “hot spots.” Deposits formed in this larger-diameter mixing zone were analyzed with scanning electron microscopy (SEM) to determine the inorganic chemistry of the ash holding the individual bed particles together. This allows a better understanding of the glue chemistry such as low melting eutectics that will limit the bed temperature to prevent these sticky ash coatings from developing (9). Tables 43 through 47 summarize the SEM morphology data from the points analyzed in the SEM photomicrographs shown in Figures 23 through 43. These figures show the location of the points analyzed and exclude any points that were mostly bed material particles (i.e., high silica). These deposits were all formed utilizing the lower velocity mixing zone.

Deposition during these tests did appear to be more prevalent with the SUFCo coal (Test P060) than the previous test (Test P057), probably as a result of the larger-diameter mixing zone resulting in poorer gas–solid mixing. The ash chemistry of the sticky phase identified in the SEM photomicrographs (see Table 43 and Figures 25–27) are similar to the low melting point calcium-iron aluminosilicate phases identified in Test P057.

The petroleum coke gasification test (Test P061) had some deposition in the burner throat area, but this deposition was not enough to prevent TRDU operation. This deposit was highly sintered (see Table 44 and Figures 28–30) and was very hard. The sticky material appeared to be derived from partially sulfided calcium and magnesium compounds and unlike the combustion ash did not appear to contain any significant levels of vanadium. Whether the deposition would have

Table 43. SEM Morphology Analysis of TRDU Mixing Zone Deposit, SUFCo Coal, Test P060 (Figures 25–27)

SEM Point No.:	1	2	3	5	6	7	8	9	10	11
Description:	Neck	Neck	Neck	Fill	Fill	Neck	Fill	Fill		Fill
	Normal O ₂ -Free Element, wt%									
Na	2.3	1.7	2.9	2.9	2.2	4.2	1.5	4.2	3.1	3.4
Mg	1.8	2.9	0.9	4.5	10.0	2.6	3.2	3.8	1.3	0.5
Al	1.8	1.4	3.7	5.7	5.2	10.3	2.9	5.7	5.9	5.5
Si	33.5	34.4	35.0	58.4	55.4	58.6	50.9	62.8	62.3	60.2
P	0.1	0.1	0.2	0.2	0.1	0.5	0.3	0.2	0.0	0.0
S	0.1	0.1	0.5	0.1	0.2	0.2	0.7	0.2	0.0	0.5
Cl	0.2	0.2	0.0	0.0	0.2	0.2	0.4	0.1	0.2	0.1
K	1.1	0.7	3.0	0.5	0.2	0.8	0.2	0.5	0.6	0.5
Ca	4.7	4.9	2.3	18.8	21.2	17.6	26.8	15.7	18.7	23.3
Ti	0.1	0.1	0.8	0.5	0.1	0.3	0.1	0.8	0.7	1.0
Cr	0.0	0.2	0.2	0.1	0.0	0.2	0.0	0.1	0.0	0.0
Fe	54.2	52.5	50.5	6.8	4.4	4.4	12.2	6.0	5.5	4.7
Ba	0.0	0.6	0.2	1.0	0.6	0.0	0.7	0.0	1.3	0.3
V	0.2	0.0	0.0	0.1	0.3	0.0	0.1	0.0	0.3	0.1
Ni	0.1	0.2	0.0	0.5	0.0	0.1	0.0	0.0	0.3	0.0
Total	100	100	100	100	100	100	100	100	100	100

Table 44. SEM Morphology Analysis of TRDU Burner Deposit, Tuscaloosa Petroleum Coke, Test P061 (Figures 28–30)

SEM Point No.:	1	2	3	4	5	6	7	9	10	11	12	13
Description:	Fill	Fill	Fill	Part.	Part.	Fill	Fill	Part.	Fill	Part.	Band	Part.
Normal O ₂ -Free Element, wt%												
Na	0.1	0.0	0.0	0.1	0.0	0.3	0.2	0.1	0.1	0.0	0.3	0.2
Mg	13.2	11.5	15.7	13.7	13.2	12.6	15.3	5.2	13.2	1.9	9.9	14.2
Al	2.5	3.8	1.8	1.5	1.4	3.6	2.0	0.6	1.5	0.1	0.8	1.4
Si	48.5	51.0	46.6	50.2	46.3	50.8	46.7	77.3	45.3	5.5	29.0	44.7
P	0.2	0.0	0.0	0.1	0.1	0.0	0.2	0.3	0.1	0.2	0.0	0.1
S	5.9	3.6	5.4	3.7	8.5	3.8	4.8	3.2	7.4	39.6	19.9	7.8
Cl	0.0	0.0	0.1	0.0	0.1	0.1	0.1	0.1	0.0	0.1	0.0	0.0
K	0.3	0.4	0.2	0.3	0.0	0.4	0.0	0.1	0.2	0.2	0.0	0.2
Ca	28.9	28.7	29.6	29.6	29.6	26.9	30.3	11.8	30.9	50.5	38.7	28.4
Ti	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.1	0.2	0.0	0.0	0.0
Cr	0.0	0.0	0.1	0.0	0.3	0.0	0.0	0.4	0.1	0.1	0.0	0.0
Fe	0.1	0.6	0.3	0.5	0.6	0.5	0.1	0.0	1.0	1.6	0.7	2.2
Ba	0.0	0.0	0.1	0.0	0.0	0.0	0.0	0.7	0.0	0.3	0.2	0.0
V	0.5	0.2	0.1	0.0	0.0	0.5	0.3	0.0	0.0	0.0	0.5	0.3
Ni	0.0	0.2	0.0	0.4	0.0	0.6	0.0	0.2	0.0	0.0	0.0	0.3
Total	100	100	100	100	100	100	100	100	100	100	100	100

Table 45. SEM Morphology Analysis of TRDU Mixing Zone Outer Deposit, Calumet Coal, Test P062 (Figures 31–33)

SEM Point No.:	1	2	3	4	5	6	7	9	10
Description:	Layer	Neck	Fill	Fill	Fill	Fill	Layer	Fill	Fill
Normal O ₂ -Free Element, wt%									
Na	0.1	0.5	0.7	0.2	0.5	0.1	0.5	0.3	0.0
Mg	0.7	0.9	1.3	0.2	0.7	0.1	0.2	0.3	0.5
Al	34.8	8.4	15.4	4.6	11.7	5.0	18.5	2.3	5.4
Si	50.5	78.1	41.7	85.5	65.2	87.1	55.8	92.3	87.1
P	0.5	0.6	0.3	0.0	0.4	0.7	0.3	0.3	0.0
S	0.0	0.0	0.0	0.1	0.3	0.2	0.0	0.0	0.2
Cl	0.2	0.0	0.0	0.0	0.1	0.0	0.1	0.0	0.2
K	4.1	5.4	13.7	3.8	8.4	2.7	18.8	2.5	1.2
Ca	0.9	1.3	0.8	0.0	2.1	0.5	0.5	0.2	1.2
Ti	1.6	0.8	2.2	0.6	2.0	0.2	0.4	0.0	0.6
Mn	0.1	0.4	0.0	0.3	0.3	0.0	0.0	0.3	0.5
Fe	5.5	3.5	24.0	3.1	8.4	2.8	3.9	0.7	3.2
Ba	1.1	0.0	0.0	1.8	0.0	0.6	1.3	0.7	0.0
Total	100.1	99.9	100.1	100.2	100	100	100	99.9	100.1

Table 46. SEM Morphology Analysis of TRDU Mixing Zone Inner Deposit, Calumet Coal, Test P062 (Figures 34–36)

SEM Point No.:	1	3	5	6	7	9	10	12
Description:	Layer	Fill	Fill	Fill	Layer	Fill	Fill	Fill
Normal O ₂ -Free Element, wt%								
Na	0.6	0.4	0.6	0.1	0.4	0.4	0.4	0.5
Mg	0.5	1.1	0.4	0.3	1.0	0.5	2.9	1.0
Al	19.0	22.2	15.2	14.5	20.2	10.0	13.5	24.6
Si	58.7	62.1	69.8	72.6	64.1	79.7	63.6	58.4
P	0.5	0.8	0.3	0.3	0.6	0.1	1.0	0.2
S	0.0	0.2	0.0	0.3	0.0	0.0	0.0	0.2
Cl	0.3	0.0	0.1	0.0	0.0	0.3	0.1	0.1
K	12.3	4.7	8.9	7.2	6.0	4.9	4.2	4.2
Ca	0.5	0.9	0.7	0.0	0.5	0.1	7.4	0.5
Ti	0.8	1.4	0.8	0.8	1.6	0.3	1.0	1.3
Mn	0.3	0.0	0.1	0.1	0.0	0.0	0.0	0.0
Fe	6.0	5.9	3.2	3.7	4.8	2.2	5.9	8.5
Ba	0.6	0.4	0.0	0.0	0.8	1.4	0.0	0.6
Total	100.1	100.1	100.1	99.9	100	99.9	100	100.1

Table 47. SEM Morphology Analysis of TRDU Mixing Zone Deposit, Illinois No. 6 Coal, Test P063 (Figures 37 and 38)

SEM Point No.:	XRFA	1	2	3	4	5	6	7	8	10	11	12
Description:	Bulk	Layer	Neck	Fill	Fill	Fill	Fill	Layer	Fill	Fill	Fill	Fill
Normal O ₂ -Free Element, wt%												
Na	0.0	0.1	0	0.1	0.0	0.0	0.0	0.0	0.1	0.0	0.3	0.0
Mg	18.7	17.0	12.0	10.7	13.6	9.0	13.9	0.7	9.1	14.3	3.0	19.7
Al	2.6	0.2	0.0	1.5	0.1	0.1	2.0	0.0	0.0	1.3	0.7	0.3
Si	26.2	0.4	0.2	33.5	0.3	0.5	24.4	0.0	4.7	3.6	7.8	6.3
P	0.1	0.2	0.0	0.1	0.2	0.4	0.4	0.2	0.0	0.3	0.1	0.0
S	9.3	32.3	37.4	0.1	33.2	37.3	16.5	42.9	38.4	33.0	30.9	32.1
Cl	ND ¹	0.0	0.0	0.0	0.0	0.4	1.4	0.0	0.0	0.2	0.0	0.1
K	0.5	0.0	0.0	0.4	0.0	0.0	0.5	0.0	0.1	0.0	0.1	0.1
Ca	33.8	44.0	47.8	53.3	21.7	49.9	31.3	41.8	45.6	19.4	11.3	36.2
Fe	8.5	3.0	2.7	0.4	30.5	1.9	9.5	14.4	2.1	27.7	45.9	5.2
Ba	ND	0.5	0.0	0.0	0.4	0.0	0.0	0.0	0.0	0.0	0.0	0.0
Ti	0.2	2.4	0.0	0.0	0.0	0.5	0.0	0.0	0.0	0.1	0.0	0.0
Total	99.9	100.1	100.1	100.1	100	100	99.9	100	100	99.9	100	100

¹ Not determined.

Table 47 (continued) (Figures 39 and 40).

SEM Point No.:	13	14	15	16	17	18	19	20	21	22	23	24	25
Description:	Layer	Neck	Fill	Fill	Fill	Fill	Layer	Fill	Fill	Fill	Fill	Fill	Layer
Normal O ₂ -Free Element, wt%													
Na	0.0	0.0	0.2	0.1	0.0	0.1	0.0	0.3	0.0	0.1	0.4	0.2	0.2
Mg	0.0	0.0	0.2	11.1	14.0	0.3	8.8	0.5	0.0	13.5	0.0	0.4	0.2
Al	0.0	0.0	0.2	0.2	2.8	0.1	0.5	0.0	0.0	1.5	0.0	0.0	0.0
Si	0.4	0.1	0.6	0.4	0.2	0.3	0.4	0.2	0.2	1.6	0.1	0.1	0.0
P	0.1	0.1	0.0	0.0	0.0	0.0	0.1	0.2	0.0	0.0	0.0	0.0	0.0
S	37.4	38.4	40.8	37.4	36.3	39.1	37.8	42.6	41.7	35.2	37.2	39.5	38.2
Cl	0.2	0.0	0.0	0.0	0.0	0.0	0.2	0.2	0.2	0.1	0.0	0.0	0.0
K	0.0	0.1	15.1	0.3	0.0	0.0	0.2	0.0	0.0	0.2	0.2	0.0	0.3
Ca	16.5	0.4	16.6	37.1	44.8	7.9	36.9	45.7	0.8	44.9	0.2	38.9	0.2
Fe	44.8	60.9	26.2	13.3	1.5	51.6	14.9	10.1	57.1	2.0	61.2	20.9	61.1
Ba	0.2	0.0	0.0	0.0	0.0	0.5	0.0	0.0	0.0	0.6	0.4	0.0	0.0
Ti	0.4	0.1	0.0	0.1	0.5	0.0	0.3	0.2	0.0	0.3	0.2	0.0	0.0
Total	99.9	100	99.9	100	100	99.9	100	100	100	100	99.9	100	100

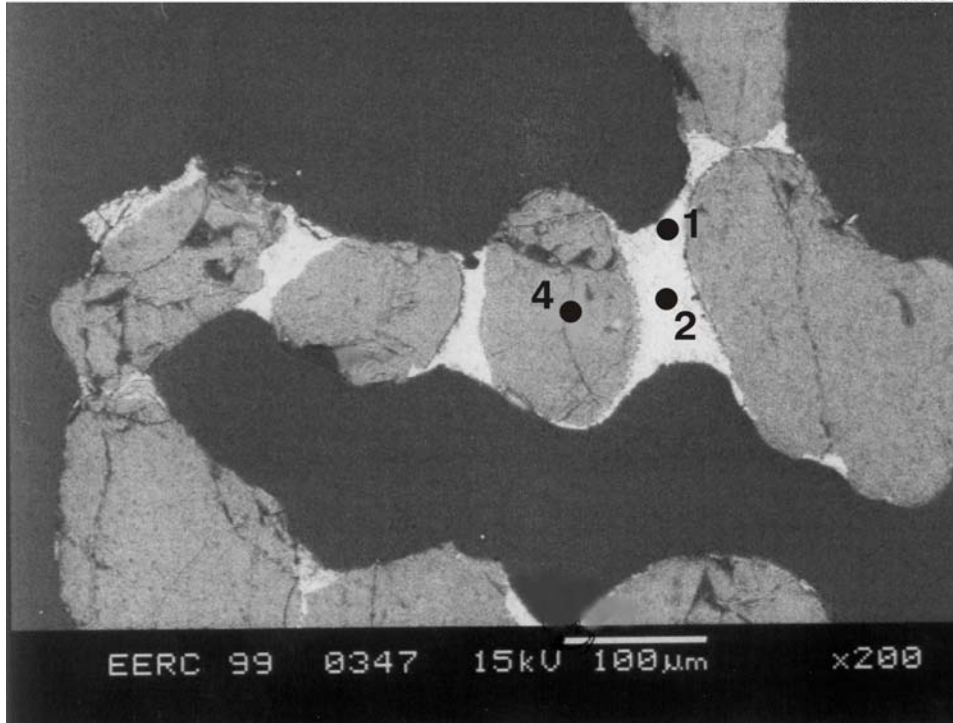


Figure 23. SEM micrograph of mixing zone deposit from gasification Test P060 on SUFCo coal showing Points 1-4.

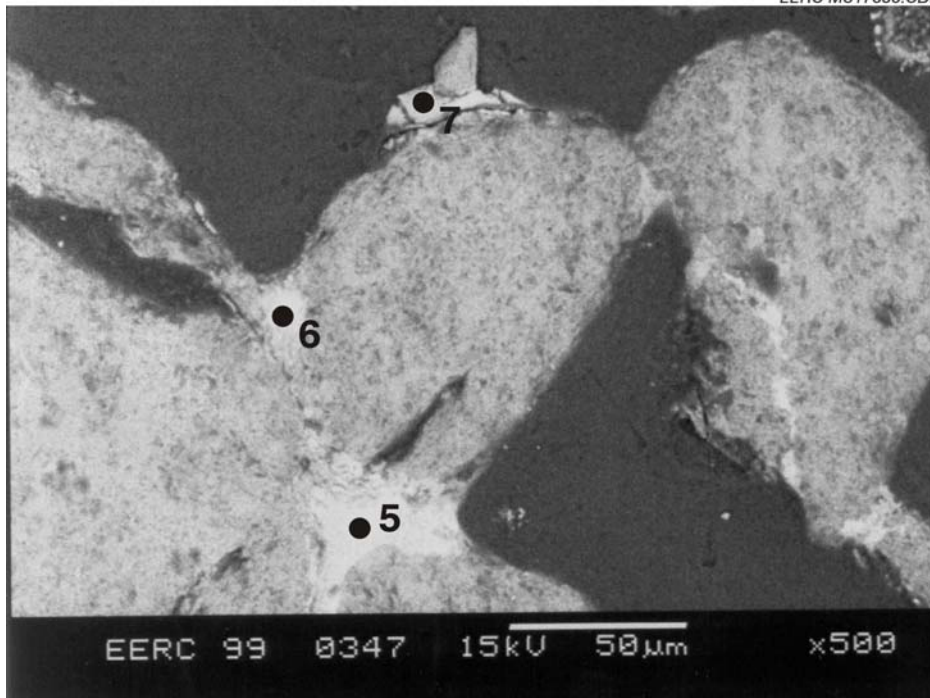


Figure 24. SEM micrograph of mixing zone deposit from gasification Test P060 on SUFCo coal showing Points 5-7.

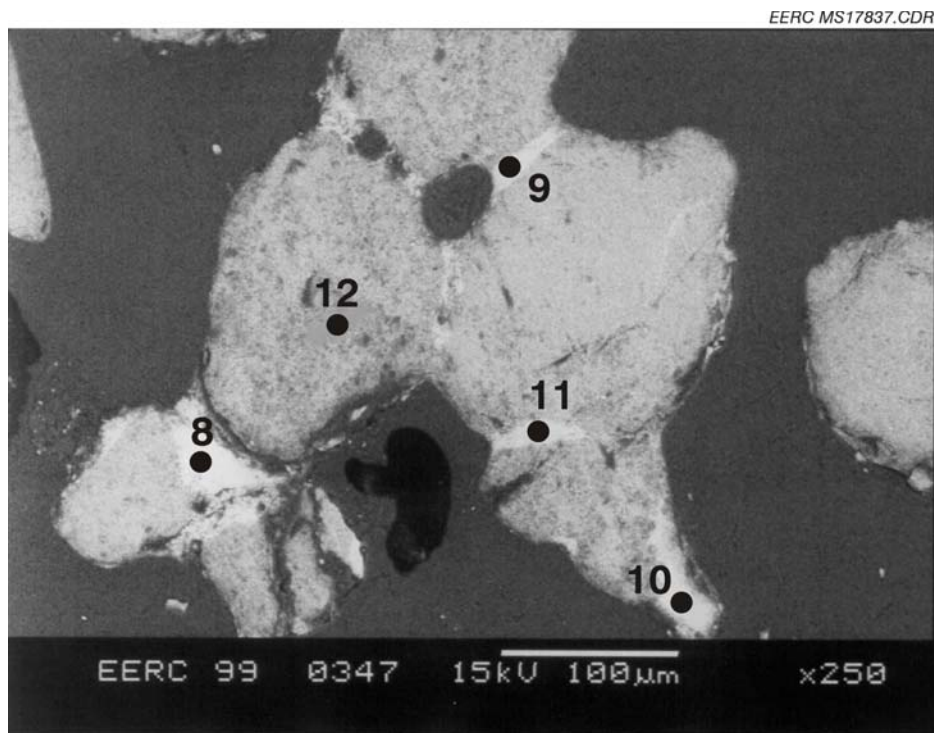


Figure 25. SEM micrograph of mixing zone deposit from gasification Test P060 on SUFCo coal showing Points 8–12.

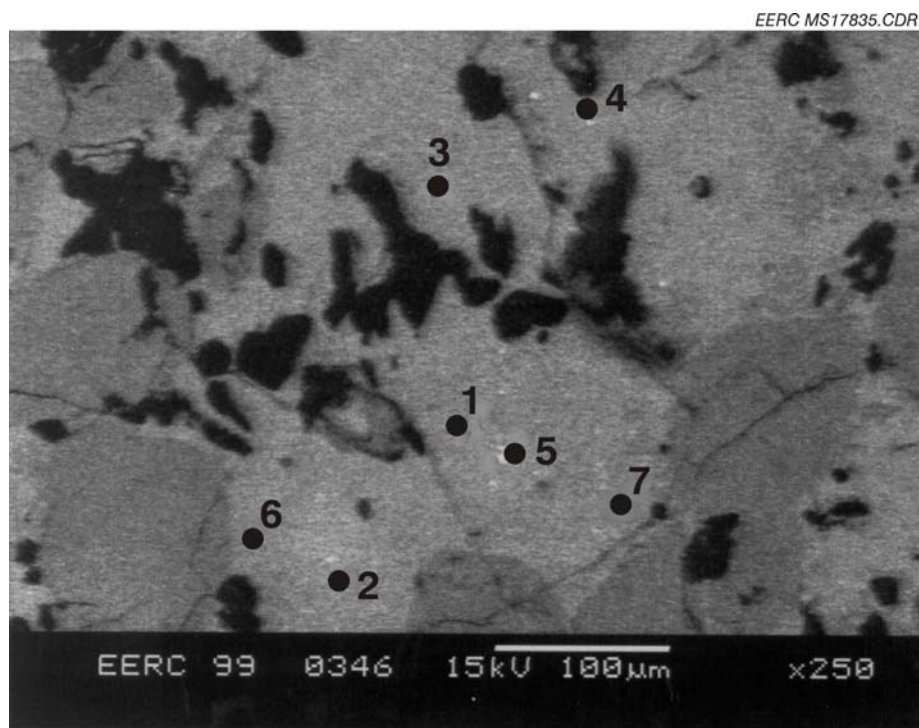


Figure 26. SEM micrograph of mixing zone deposit from gasification Test P061 on Tuscaloosa petcoke showing Points 1–7.

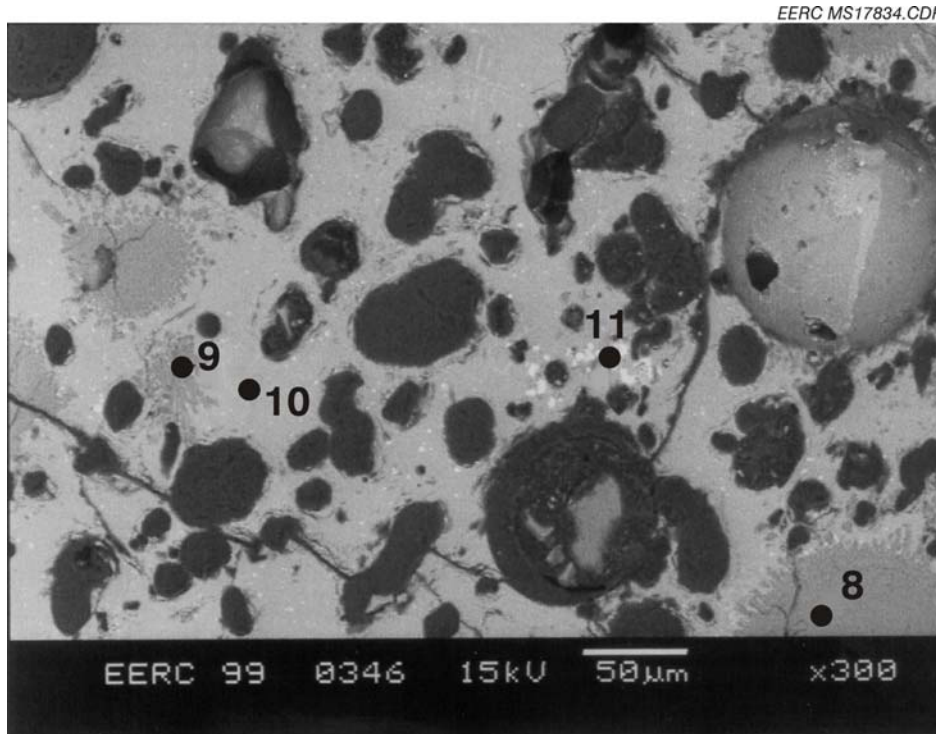


Figure 27. SEM micrograph of mixing zone deposit from gasification Test P061 on Tuscaloosa showing Points 8–11.



Figure 28. SEM micrograph of mixing zone deposit from gasification Test P061 on Tuscaloosa petcoke showing Points 12–14.

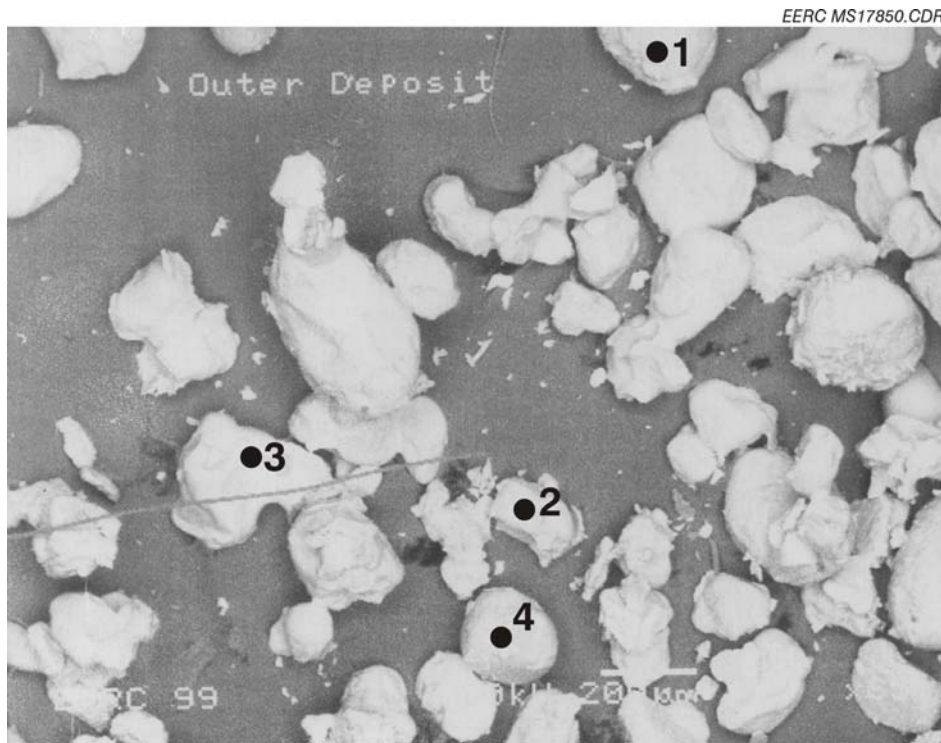


Figure 29. SEM micrograph of outer mixing zone deposit from gasification Test P062 on Calumet Mine bituminous coal showing Points 1–4.

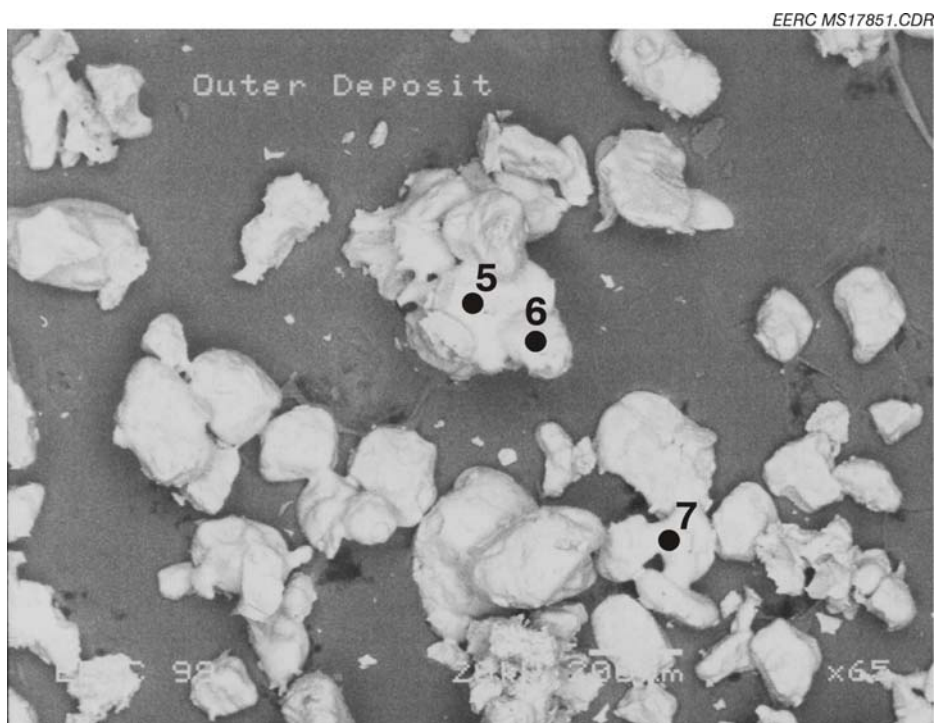


Figure 30. SEM micrograph of outer mixing zone deposit from gasification Test P062 on Calumet Mine bituminous coal showing Points 5–7.

continued to the point that it did interrupt steady TRDU operation or reached some maximum level where the increased burner velocity would keep the remaining throat opening clear is not known.

The short gasification test (Test P062) with the Alabama bituminous coal from Calumet Mine had substantial deposition in the mixing zone which was also compounded by the buildup of char agglomerates there. These agglomerates were the result of feeding coal that had more swelling properties than originally expected.

The buildup of these agglomerates in the mixing zone was leading to poor gas–solid mixing and probably to the presence of localized hot spots in the mixing zone. The SEM morphology analysis (Tables 45 and 46 and Figures 31–36) indicates that the melted sticky phase in the deposits was due to a low melting point potassium aluminosilicate, which was generated from the high levels of illite found in the starting coal. Illite has been shown to form low melting point compounds, especially under reducing conditions (10). No significant levels of sulfur were found in the deposit either at the oxidized inner layer or the carbon-rich outer layer, thereby suggesting that the formation of a low melting point sulfide compound was probably not the mechanism for deposit formation.

The Illinois No. 6 gasification test (Test P063) was operated successfully under air-blown operating conditions and the first attempt with oxygen-enriched air testing (26 mol%). It was not until a higher level of enriched air testing (33 mol%) was started that operating temperatures rose to over 1100°C (2012°F) and deposits formed very rapidly and terminated the test. Table 47 and Figures 37–40 show the SEM morphology of the mixing zone deposits. These deposits were very sintered and quite hard with low porosity because of a large portion of the deposit material melting. The deposit chemistry is very similar to the chemistry seen in the previous Illinois No. 6 gasification test (Test P056) in that a very high concentration of iron sulfide was observed in the deposit. During this test, it was also observed that even a small increase in the solid circulation rate could decrease the operating temperature by 30°C (86°F), suggesting that modifications to the TRDU that substantially increase circulation rate will allow the reactor temperature to be controlled even at higher oxygen levels.

Ash behavior in power systems can have a significant impact on the design and performance of advanced power systems. The EERC has focused significant effort on ash behavior in conventional power systems that can be applied to advanced power systems. This program utilized methods developed to better understand and mitigate adverse coal ash behavior in the EERC TRDU; however, these methods and observations can also be applied to other advanced power systems.

Although it is well established that sulfides readily break down in combustion environments, usually into oxides of iron, the mechanisms of sulfide formation in gasification are not well understood. Work by Benson and Sondreal (10) revealed that initial sulfidation of coal ash or bed material may have led to Ca–Mg-rich aluminosilicate deposits that formed in a pressurized circulating fluidized-bed gasifier. Volatile sulfide species can exist in the temperatures noted for the gasifier studied by Benson and Sondreal (10) and also in other gasification environments. Low melting point sulfides of Ca, Fe or, possibly, even Na are stable at temperatures less than 900°C (1652°F) in these environments, but the specific interaction of how the sulfides could lead to other silicate and oxide components becoming the “glue” material in a deposit is not understood.

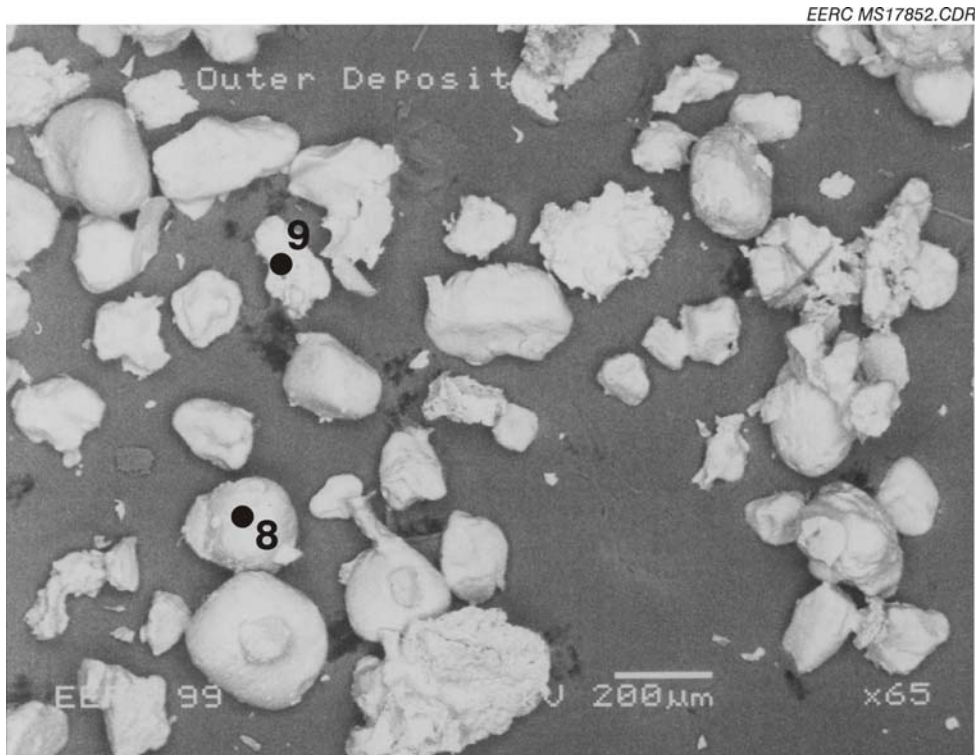


Figure 31. SEM micrograph of outer mixing zone deposit from gasification Test P062 on Calumet Mine bituminous coal showing Points 8 and 9.

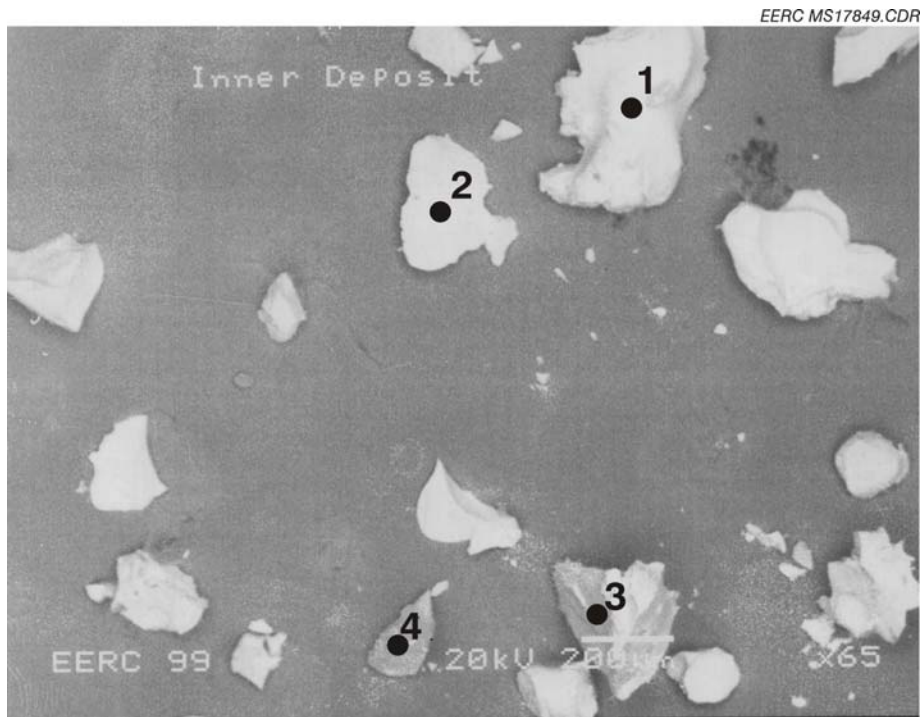


Figure 32. SEM micrograph of inner mixing zone deposit from gasification Test P062 on Calumet Mine bituminous coal showing Points 1-4.

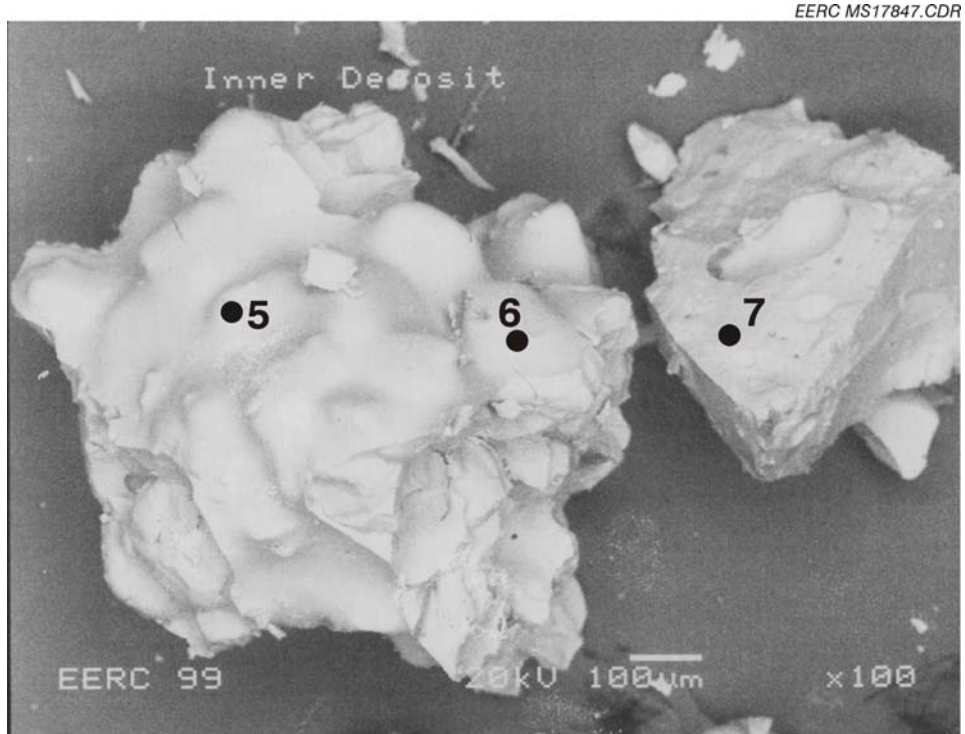


Figure 33. SEM micrograph of inner mixing zone deposit from gasification Test P062 on Calumet Mine bituminous coal showing Points 5-7.

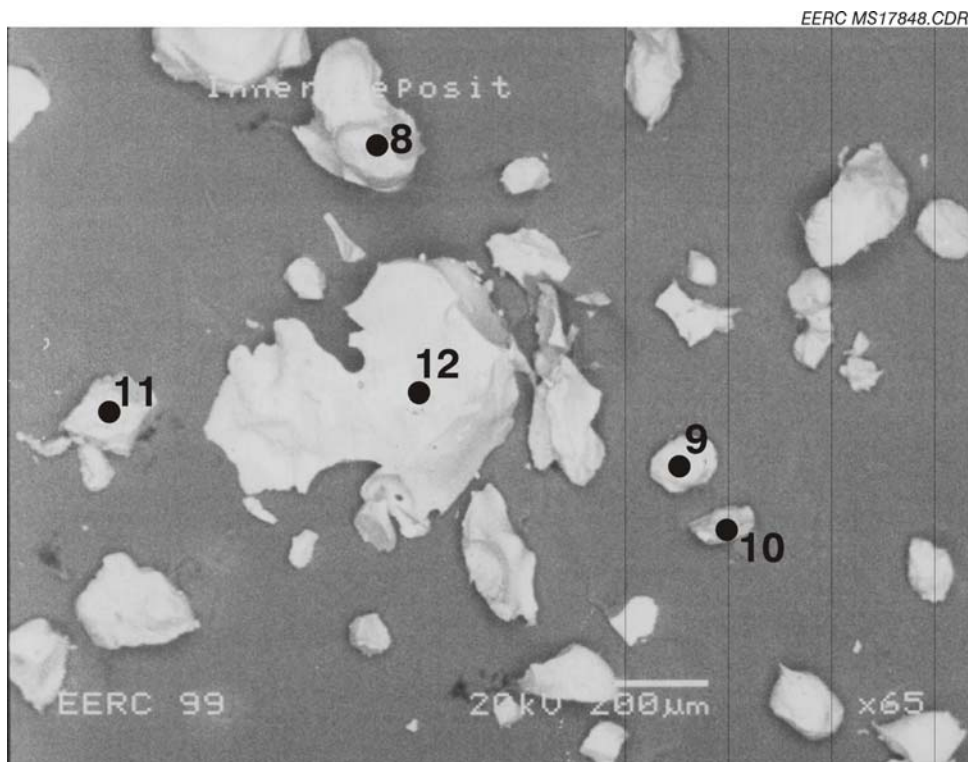


Figure 34. SEM micrograph of inner mixing zone deposit from gasification Test P062 on Calumet Mine bituminous coal showing Points 8-12.

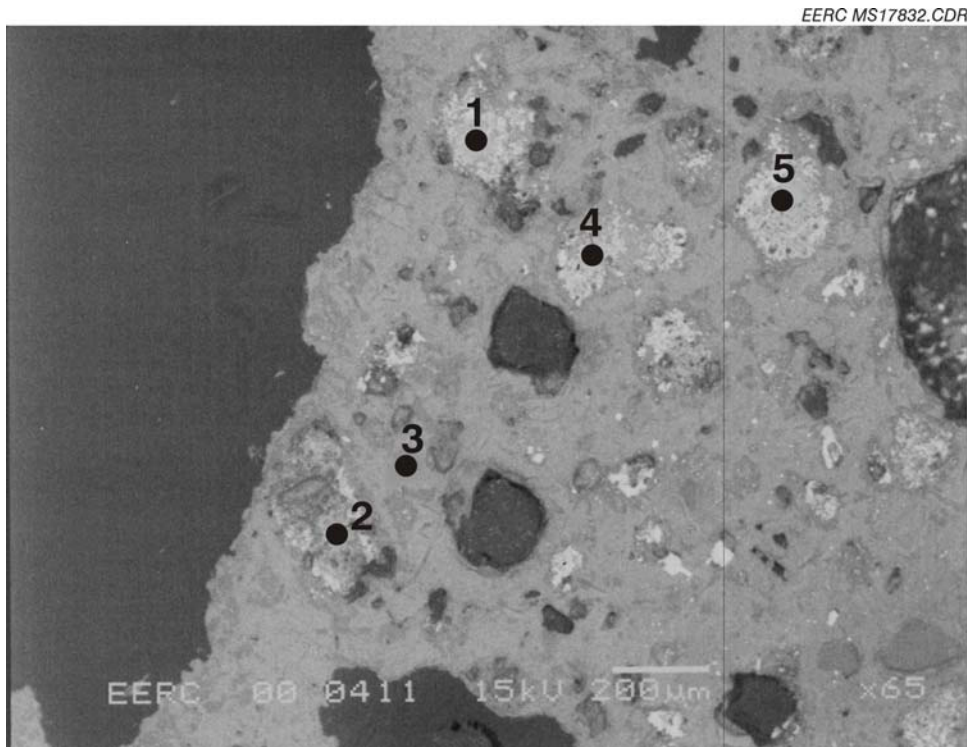


Figure 35. SEM micrograph of mixing zone deposit from gasification Test P063 on Illinois No. 6 bituminous coal showing Points 1–5.

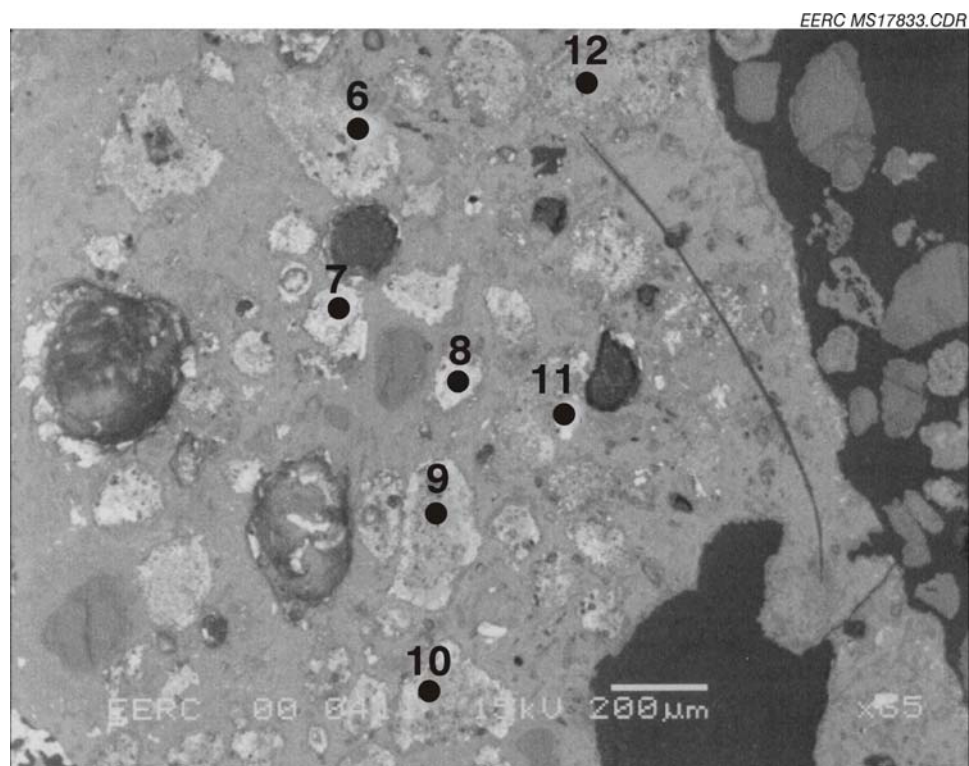


Figure 36. SEM micrograph of mixing zone deposit from gasification Test P063 on Illinois No. 6 bituminous coal showing Points 6–12.

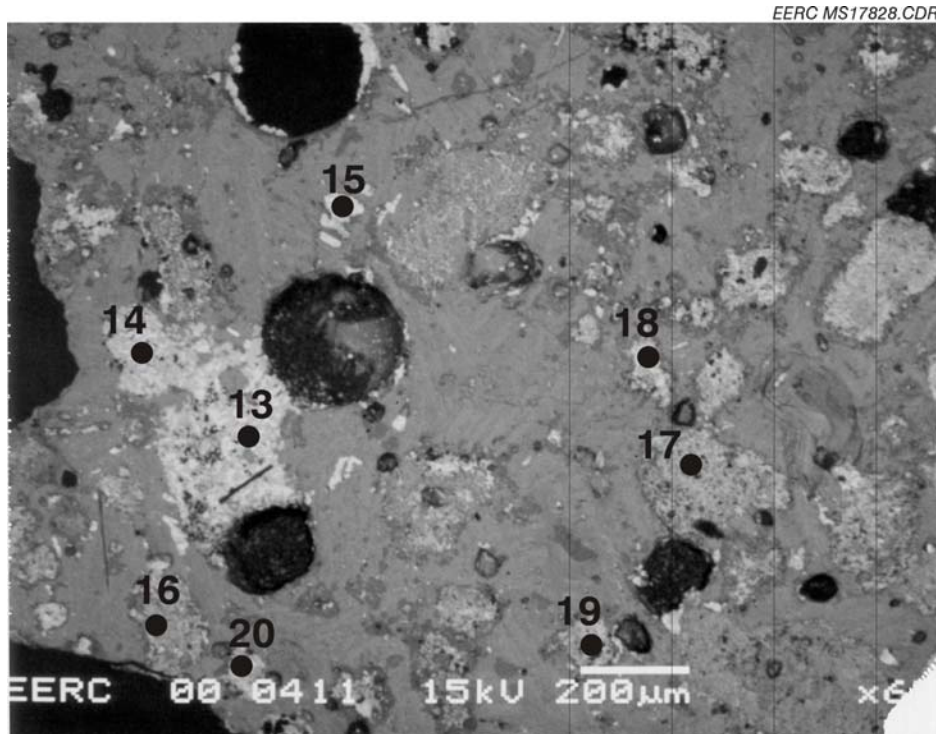


Figure 37. SEM micrograph of mixing zone deposit from gasification Test P063 on Illinois No. 6 bituminous coal showing Points 13–20.

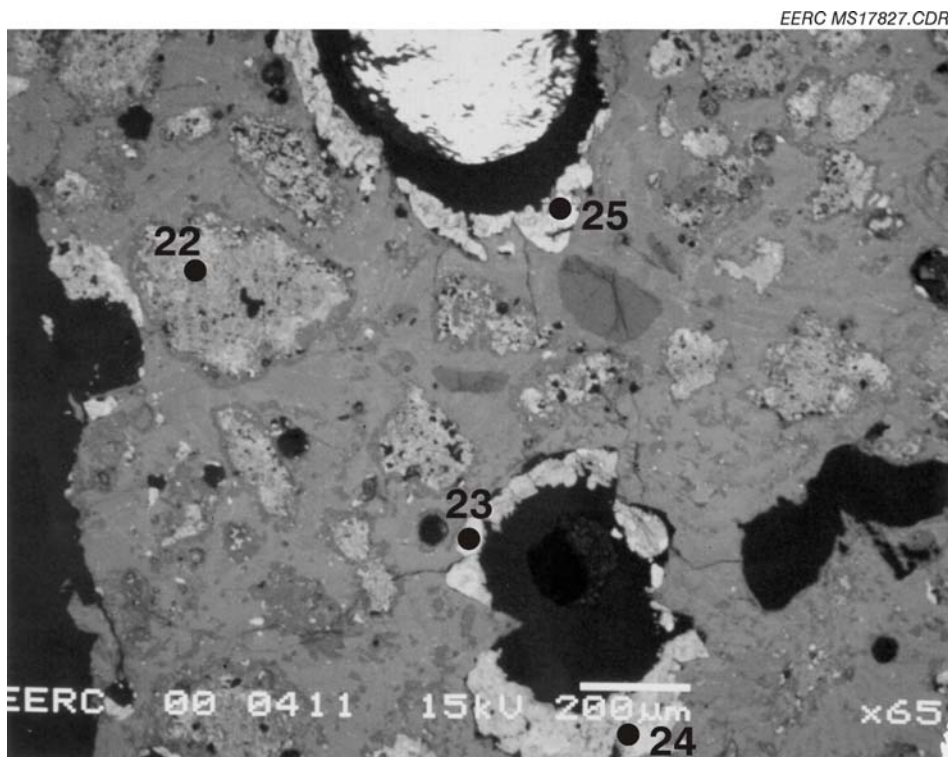


Figure 38. SEM micrograph of mixing zone deposit from gasification Test P063 on Illinois No. 6 bituminous coal showing Points 22–25.

Table 48. Maximum Reactive Sulfide Levels Determined for Various TRDU Samples

Sample Description	LASH Bed Material, $\mu\text{g/g}$	Filter Ash, $\mu\text{g/g}$
P067 Navajo	134	0.68
P071 Petcoke	42.8	<0.5
P071 Prater Creek	56.8	<0.5
P072 Illinois No. 6	1.5	<0.2
P073 Blacksville (Pitts. 8)	30	<0.2
P074 Falkirk–Wood	5.26	<0.4
P075 Loy Yang	<0.3	<0.5
P075 Lochiel	4.69	<0.4

Ash deposits were collected during runs of the TRDU gasifier, and the mechanisms of ash deposition were assessed, initially with a specific view toward the role of sulfides. Process information and deposits from the TRDU have provided deposits and process data with which to propose some deposition mechanisms. The interactions between sulfides and silicate or oxide materials can lead to potentially serious ash deposit formation, thereby adversely affecting system performance.

4.8 Determination of Reactive Sulfides in TRDU Samples

One potential issue for the utilization of calcium-based sorbents under reducing conditions is the formation of reactive sulfides in the solid materials removed from the gasifier. Reactive sulfide levels above $500 \mu\text{g/g}$, as determined by EPA 376.2, are considered a hazardous waste that either must be disposed of or combusted to convert the sulfide species to a sulfate. North Dakota regulations require that hazardous wastes that are burned for energy recovery have a heating value of greater than 5000 Btu/lb or the energy recovery process can be considered “sham” recycling. The amount of carbon present in the limestone ash (LASH) bed material for all fuels tested (generally less than 20 wt% and in most cases approximately 10 wt%) indicate that the LASH by itself would not be recyclable because of the low heating value. The high carbon content of the filter ash material, typically greater than 40 wt%, suggests that this material would have enough heating value for recycling. Recent economic studies performed by Southern Company Services (SCS) on the transport reactor gasifier indicate that the most economic disposal option would be to landfill the ash without any treatment. However, should reactive sulfide levels limit this disposal, both the LASH bed material and the filter ash could be recycled to a fluid-bed combustor in an integrated commercial system. Mixing these fuels should provide a recycled stream that meets the 5000-Btu/lb requirement. Reactive sulfide tests were performed on various samples obtained from the TRDU while using the three different fuels tested under gasification conditions. Table 48 shows the reactive sulfide levels determined for these streams.

From these analyses, it appears that as the particle size gets smaller, less reactive sulfide is present in the solid material; thus no filter ash was measured with reactive sulfide levels above even $1 \mu\text{g/g}$. The lower sulfur fuels provided LASH material that was still well below the allowable 500

μg/g reactive sulfide levels that make the material a hazardous waste, and even the high-sulfur fuels generated bed material that was still below the allowable 500 μg/g reactive sulfide limits. Other work performed at the EERC has shown that the combustion of these types of materials has been very successful in converting the reactive sulfide species to sulfates, thereby rendering them inert (see Appendix B).

4.9 TRDU Sulfur Capture Performance

The TRDU has shown a marked decrease in sulfur capture when the transport reactor has been operated in oxygen-blown mode. The H₂S concentration has ranged from 1200 to as high as 9000 ppm under full oxygen-blown operating conditions. The sulfur retention has ranged from 15% to 40% for the lower-sulfur coals and as high as 50% to 60% for the high-sulfur bituminous coals. This relatively low level of sulfur capture is the result of the high water and carbon dioxide partial pressures generated by oxygen-blown operation greatly reducing the equilibrium concentration of calcium sulfide that will form according to the reaction:



Sulfur retention data are given in the individual steady-state data presented in Appendix A.

5.0 CONCLUSIONS

The TRDU was modified to accommodate oxygen-blown operation in support of a Vision 21-type energy plex which could produce power, chemicals, and fuel. These modifications consisted of changing the loop seal design from a J-leg to an L-valve configuration, thereby increasing the mixing zone length and residence time. In addition, the standpipe, dipleg, and L-valve diameters were increased to reduce slugging caused by bubble formation in the lightly fluidized sections of the solid return legs. A seal pot was added to the bottom of the dipleg so that the level of solids in the standpipe could be operated independently of the dipleg return leg. A separate coal feed nozzle was added that could inject the coal upward into the outlet of the mixing zone, thereby precluding any chance of the fresh coal feed back-mixing into the oxidizing zone of the mixing zone; however, difficulties with this coal feed configuration led to a switch back to the original downward configuration. Instrumentation to measure and control the flow of oxygen and steam to the burner and mix zone ports was added to allow the TRDU to be operated under full oxygen-blown conditions.

In total, ten test campaigns have been conducted under enriched air- or full oxygen-blown conditions. During these tests, 1515 hours of coal feed with 660 hours of air-blown gasification and 720 hours of enriched air- or oxygen-blown coal gasification were completed. During these tests, approximately 366 hours of operation with Wyodak, 123 hours of operation with Navajo subbituminous coal, 143 hours of operation on Illinois No. 6, 106 hours on SUFCo, 110 hours on Prater Creek, 48 hours on Calumet, and 134 hours on a Pittsburgh No. 8 bituminous coals were completed. In addition, 331 hours of operation on low-rank coals such as North Dakota lignite, Australian brown coal, and a 90:10 wt% mixture of lignite and wood waste were completed. Also included in these test campaigns was 50 hours of gasification on a petroleum coke from the Hunt

Oil Refinery, and an additional 73 hours of operation on a high ash coal from India was completed. Data from these tests indicate that the transport gasifier performs better on the lower-rank feedstocks because of their higher char reactivity with the gasification reactions.

Comparable carbon conversions have been achieved at similar oxygen/coal ratios for both air-blown and oxygen-blown operation. While separation of fines from the feed coals is not needed with this technology, some testing has suggested that feedstocks with high levels of fines have resulted in reduced performance. These data show that these low-rank feedstocks provided similar fuel gas heating values; however, even among the high-reactivity low-rank coals, the carbon conversion did appear to lower for the fuels (brown coal in particular) that contained a significant amount of fines. The fuel gas under oxygen-blown operation has been high in hydrogen and carbon dioxide concentration since the high steam injection rate drives the water-gas shift reaction to produce more CO₂ and H₂ at the expense of the CO and water vapor. However, the high water and CO₂ partial pressures have also greatly retarded the reaction of hydrogen sulfide with the calcium-based sorbents.

Since warm gas cleanup is utilized, the unconverted steam and coal moisture injected into the gasifier will remain in the fuel gas entering the gas turbine. When the air-blown and oxygen-blown fuel gas heating values are compared for the wet product gas streams, it is apparent that only a slight improvement in product gas heating value entering the gas turbine is achieved with oxygen-blown operation. In order to keep the gas turbine firing temperature down to prevent thermal NO_x formation, typically large amounts of nitrogen or steam are injected into the gas turbine combustor such that the fuel gas heating is typically not much greater than 115 Btu/scf as-fired. In essence, the transport reactor has either injected the nitrogen with the oxidant (in the form of air) into the gasifier instead of directly into the gas turbine combustor in air-blown mode or has injected the steam directly into the gasifier instead of the gas turbine combustor in the oxygen-blown case. However, in a Vision 21 plant, where chemicals or fuel production are being considered and where potentially conventional cold-gas cleanup technology would be utilized to remove the water vapor from the fuel gas stream, significantly higher concentrations of desirable fuel gas constituents are achieved with oxygen-blown operation.

The TRDU and hot-gas filters have operated for over 2175 hours in gasification mode and over 2500 hours total with no major candle failures. The candles have exhibited no significant loss in candle permeability. The baseline “cleaned” filter differential pressure typically increased from 20 to approximately 80 inches H₂O over the course of most tests. The inlet particulate loading has ranged from approximately 3500 to 33,800 ppm with the filter ash averaging between 20 to 70 wt% carbon with a low bulk density around 20 lb/ft³. The average filter ash particle size has ranged from approximately 7 to 22 μm in size and was essentially representative of the coal ash from very early in the gasification test. The initial rapid recovery of the filter differential pressure along with the small size, the lack of the cohesiveness seen in other filter ashes, and the low density of the ash had suggested that a high percentage of the filter cake would be reentrained back onto the filters after they are backpulsed. The large increase in filter baseline differential pressure also suggests that a thin but low porosity (permeable) filter cake is remaining on the surface of the candle and is not being removed during backpulsing. The low bulk density and high flowability of the filter ash possibly suggests that the inlet ash is able to move or sift on the surface of the candle to reach some optimum (minimum) porosity leading to low gas permeability across the candle.

Continuous measurement of mercury in the warm fuel gas has been another goal of the project. After considerable trial and error, a fuel gas conditioning system and Hg CEM analyzer has been configured to allow the continuous measurement of mercury emissions. Sampling issues for both the wet chemistry and Hg CEM techniques have been resolved, so that good agreement between the two techniques is being achieved. Wet-chemistry analysis has shown the mercury to essentially be in the elemental form. The EERC continues to utilize advanced SEM techniques where appropriate to determine the chemistry of any bed material agglomeration or deposition samples. No high levels of reactive sulfide have been measured in any TRDU samples that would make the residual solids a hazardous waste.

6.0 FUTURE PLANS

Future plans for operation of the TRDU include the design and construction of a new gas and particulate sampler for obtaining hot samples from the mixing zone and riser of the TRDU. Other future plans include testing of potential water-gas shift catalyst and membranes in order to maximize hydrogen production from future feedstocks. This testing will focus on demonstrating the transport gasifier as a significant hydrogen producer to help fuel the potential hydrogen economy. The EERC will continue aiming to integrate TRDU testing with potential partners focused on developing lower-cost synergistic multicontaminant control devices for warm-gas applications.

7.0 REFERENCES

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APPENDIX A

TRDU OXYGEN-BLOWN OPERATIONAL DATA WITH MODIFIED L-VALVE LOOP SEAL

Gas Compositions from TRDU Test

Oxygen-blown lignite tests P068

Center-1 Center-2 Center-3 Center-4 Center-5 Center-6 Center-7 Center-8 Falkirk-1 Falkirk-2 Falkirk-3 Falkirk-4 Freedom-1 Freedom-2 Freedom-3 HS Free-1 HS Free-2 HS Free-3 HS Free-4
 06/12/2001 06/14/2001 06/14/2001 06/14/2001 06/15/2001 06/15/2001 06/16/2001 06/16/2001 06/16/2001 06/17/2001 06/17/2001 06/17/2001 06/18/2001 06/18/2001 06/19/2001 06/20/2001 06/20/2001 06/20/2001 06/21/2001 06/21/2001 06/21/2001 06/21/2001
 13:00-20:00 06:00-10:00 13:00-17:00 20:00-24:00 00:00-8:00 11:00-23:00 00:00-7:00 8:00-12:00 15:00-24:00 01:00-17:00 18:00-23:00 00:00-07:00 11:00-17:00 00:00-10:00 12:30-14:00 17:00-24:00 00:00-5:00 06:00-13:00 06:00-15:00 15:00-18:00

Product Gas Comp, vol%	Center-1	Center-2	Center-3	Center-4	Center-5	Center-6	Center-7	Center-8	Falkirk-1	Falkirk-2	Falkirk-3	Falkirk-4	Freedom-1	Freedom-2	Freedom-3	HS Free-1	HS Free-2	HS Free-3	HS Free-4
H2	8.2	6.7	6.4	13.5	12.7	18.5	19.5	20.8	19.8	18.8	18.3	19.5	9.3	8.6	16.5	17.3	20.2	19.9	20.0
CO	3.2	3.4	2.9	4.8	4.7	6.1	6.6	7.4	7.0	6.9	6.8	6.9	4.7	4.7	4.8	5.7	7.1	6.8	6.6
CH4	1.1	1.1	1.1	1.9	1.9	2.8	3.1	3.1	2.8	2.7	2.6	2.8	1.3	1.1	1.9	2.2	2.4	2.3	2.4
CO2	14	14	14	21	21	28	29	29	29	29	33	29	16	15	26	30	29	29	30
N2	73	74	76	57	57	43	41	39	40	43	44	43	73	72	50	48	43	44	44
Total	99	99	100	98	97	99	99	99	99	100	105	100	104	102	99	103	101	101	103
Ave Mol Wt	28	28	28	27	27	27	27	27	27	27	30	27	29	29	27	29	27	28	28

Flow, scfh	Center-1	Center-2	Center-3	Center-4	Center-5	Center-6	Center-7	Center-8	Falkirk-1	Falkirk-2	Falkirk-3	Falkirk-4	Freedom-1	Freedom-2	Freedom-3	HS Free-1	HS Free-2	HS Free-3	HS Free-4
air in	10542	10500	11000	4250	4364	0	0	0	0	0	0	0	10999	11000	0	0	0	0	0
oxygen in	0	0	0	1549	1549	2504	2563	2702	2570	2526	2474	2502	0	0	2412	2379	2488	2478	2449
nitrogen in	4082	3991	4676	4063	4065	4123	4123	4146	4088	4140	4133	4198	4324	4340	5380	3998	4030	4062	4002
product gas	19452	19044	20612	15627	15638	12503	13075	13768	13110	12611	12341	12920	21392	21419	13495	11623	12941	12668	12537

Product Gas, scfh	Center-1	Center-2	Center-3	Center-4	Center-5	Center-6	Center-7	Center-8	Falkirk-1	Falkirk-2	Falkirk-3	Falkirk-4	Freedom-1	Freedom-2	Freedom-3	HS Free-1	HS Free-2	HS Free-3	HS Free-4
H2	1593	1276	1311	2105	1989	2307	2554	2869	2601	2376	2263	2525	1989	1836	2228	2012	2610	2515	2502
CO	628	644	606	745	740	768	858	1017	923	875	835	894	995	1007	652	666	921	864	825
CH4	212	209	223	291	291	354	399	421	371	346	326	357	280	231	256	250	307	296	298
CO2	2723	2731	2851	3254	3261	3550	3772	3971	3782	3631	4110	3693	3329	3269	3509	3489	3701	3610	3729
N2	14155	14045	15579	8928	8920	5346	5424	5371	5291	5371	5400	5503	15623	15452	6748	5530	5581	5565	5548
Total	19312	18905	20569	15322	15200	12324	13006	13650	12968	12598	12935	12970	22216	21794	13392	11947	13121	12850	12902

Heating Value, Btu/hr	Center-1	Center-2	Center-3	Center-4	Center-5	Center-6	Center-7	Center-8	Falkirk-1	Falkirk-2	Falkirk-3	Falkirk-4	Freedom-1	Freedom-2	Freedom-3	HS Free-1	HS Free-2	HS Free-3	HS Free-4
H2 (325)	517764	414683	426050	684111	646475	749711	829903	932507	845333	772172	735585	820485	646573	596573	724108	653881	848315	817244	813275
CO (321)	201684	206624	194524	239276	237436	246427	275328	326603	296265	280940	268191	286995	319308	323148	209231	213785	295769	277330	264804
CH4 (1014)	214995	212417	225726	294731	294939	358789	404371	427199	376207	350379	330364	361584	284158	234564	259995	253393	310996	300581	302558
Total	934443	833723	846300	1218118	1178850	1354926	1509602	1686309	1517805	1403491	1334140	1469063	1250039	1154285	1193333	1121059	1455080	1395156	1380637

Heating Value, Btu/scf	Center-1	Center-2	Center-3	Center-4	Center-5	Center-6	Center-7	Center-8	Falkirk-1	Falkirk-2	Falkirk-3	Falkirk-4	Freedom-1	Freedom-2	Freedom-3	HS Free-1	HS Free-2	HS Free-3	HS Free-4
H2	27	22	21	44	41	60	63	68	64	61	60	64	30	28	54	56	66	65	65
CO	10	11	9	15	15	20	21	24	23	22	22	22	15	15	16	18	23	22	21
CH4	11	11	11	19	19	29	31	31	29	28	27	28	13	11	19	22	24	24	24
Total	48	44	41	78	75	108	115	122	116	111	108	114	58	54	88	96	112	110	110

VALUES ADJUSTED FOR PURGE NITROGEN

Product Gas Comp, vol%	Center-1	Center-2	Center-3	Center-4	Center-5	Center-6	Center-7	Center-8	Falkirk-1	Falkirk-2	Falkirk-3	Falkirk-4	Freedom-1	Freedom-2	Freedom-3	HS Free-1	HS Free-2	HS Free-3	HS Free-4
H2	10.5%	8.6%	8.2%	18.7%	17.9%	28.1%	28.7%	30.2%	29.3%	28.1%	25.7%	28.8%	11.1%	10.5%	27.8%	25.3%	28.7%	28.6%	28.1%
CO	4.1%	4.3%	3.8%	6.6%	6.6%	9.4%	9.7%	10.7%	10.4%	10.3%	9.5%	10.2%	5.6%	5.8%	8.1%	8.4%	10.1%	9.8%	9.3%
CH4	1.4%	1.4%	1.4%	2.6%	2.6%	4.3%	4.5%	4.4%	4.2%	4.1%	3.7%	4.1%	1.6%	1.3%	3.2%	3.1%	3.4%	3.4%	3.4%
CO2	17.9%	18.3%	17.9%	28.9%	29.3%	43.3%	42.5%	41.8%	42.6%	42.9%	46.7%	42.1%	18.6%	18.7%	43.8%	43.9%	40.7%	41.1%	41.9%
N2	66.1%	67.4%	68.6%	43.2%	43.6%	14.9%	14.6%	12.9%	13.5%	14.6%	14.4%	14.9%	63.2%	63.7%	17.1%	19.3%	17.1%	17.1%	17.4%
Total	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%
Ave Mol Wt	28	29	29	27	28	27	27	26	27	27	28	27	28	28	27	28	27	27	27

Heating Value, Btu/scf	Center-1	Center-2	Center-3	Center-4	Center-5	Center-6	Center-7	Center-8	Falkirk-1	Falkirk-2	Falkirk-3	Falkirk-4	Freedom-1	Freedom-2	Freedom-3	HS Free-1	HS Free-2	HS Free-3	HS Free-4
	61	56	53	108	106	165	170	177	171	166	152	167	70	66	149	141	160	159	155

Product Gas, scfh	Center-1	Center-2	Center-3	Center-4	Center-5	Center-6	Center-7	Center-8	Falkirk-1	Falkirk-2	Falkirk-3	Falkirk-4	Freedom-1	Freedom-2	Freedom-3	HS Free-1	HS Free-2	HS Free-3	HS Free-4
H2	1593	1276	1311	2105	1989	2307	2554	2869	2601	2376	2263	2525	1989	1836	2228	2012	2610	2515	2502
CO	628	644	606	745	740	768	858	1017	923	875	835	894	995	1007	652	666	921	864	825
CH4	212	209	223	291	291	354	399	421	371	346	326	357	280	231	256	250	307	296	298
CO2	2723	2731	2851	3254	3261	3550	3772	3971	3782	3631	4110	3693	3329	3269	3509	3489	3701	3610	3729
N2	10073	10054	10903	4865	4855	1223	1301	1225	1203	1231	1267	1305	11299	11112	1368	1532	1551	1503	1546
Total	15230	14914	15893	11259	11135	8201	8883	9504	8880	8458	8802	8772	17892	17454	8012	7949	9091	8788	8900

FLUE GAS FLOWS ADJUSTED TO 450,000 BTU/HR HEAT LOSS

Gas created due to combustion of	Center-1	Center-2	Center-3	Center-4	Center-5	Center-6	Center-7	Center-8	Falkirk-1	Falkirk-2	Falkirk-3	Falkirk-4	Freedom-1	Freedom-2	Freedom-3	HS Free-1	HS Free-2	HS Free-3	HS Free-4
Gas created due to combustion of	68	68	68	68	68	68	68	68	72	72	72	72	71	71	71	71	71	71	71

	Center-1	Center-2	Center-3	Center-4	Center-5	Center-6	Center-7	Center-8	Falkirk-1	Falkirk-2	Falkirk-3	Falkirk-4	Freedom-1	Freedom-2	Freedom-3	HS Free-1	HS Free-2	HS Free-3	HS Free-4
CO2	1060	1060	1060	1060	1060	1060	1060	1060	1121	1121	1121	1121	1106	1106	1106	1106	1106	1106	1106
N2	4487	4487	4487	4487	4487	4487	4487	4487	500	500	500	500	4679	4679	4679	4679	4679	4679	4679

Product Gas, scfh	Center-1	Center-2	Center-3	Center-4	Center-5	Center-6	Center-7	Center-8	Falkirk-1	Falkirk-2	Falkirk-3	Falkirk-4	Freedom-1	Freedom-2	Freedom-3	HS Free-1	HS Free-2	HS Free-3	HS Free-4
H2	1593	1276	1311	2105	1989	2307	2554	2869	2601	2376	2263	2525	1989	1836	2228	2012	2610	2515	2502
CO	628	644	606	745	740	768	858	1017	923	875	835	894	995	1007	652	666	921	864	825
CH4	212	209	223	291	291	354	399	421	371	346	326	357	280	231	256	250	307	296	298
CO2	1663	1671	1790	2193	2200														

		Oxygen-blown lignite tests P068																			
		Center-1	Center-2	Center-3	Center-4	Center-5	Center-6	Center-7	Center-8	Falkirk-1	Falkirk-2	Falkirk-3	Falkirk-4	Freedom-1	Freedom-2	Freedom-3	HS Free-1	HS Free-2	HS Free-3	HS Free-4	
		06/12/2001	06/14/2001	06/14/2001	06/14/2001	06/15/2001	06/15/2001	06/16/2001	06/16/2001	06/16/2001	06/17/2001	06/17/2001	06/18/2001	06/18/2001	06/19/2001	06/20/2001	06/20/2001	06/21/2001	06/21/2001	06/21/2001	06/21/2001
		13:00-20:00	06:00-10:00	13:00-17:00	20:00-24:00	00:00-8:00	11:00-23:00	00:00-7:00	8:00-12:00	15:00-24:00	01:00-17:00	18:00-23:00	00:00-07:00	11:00-17:00	00:00-10:00	12:30-14:00	17:00-24:00	00:00-5:00	06:00-13:00	15:00-18:00	
Product Gas, vol %																					
	H2	17	14	13	37	36	38	38	39	36	35	32	35	16	16	39	34	40	40	39	
	CO	6.5	6.9	5.9	13.1	13.2	12.7	12.9	13.8	12.7	12.8	11.6	12.5	8.2	8.6	11.5	11.2	14.0	13.7	12.9	
	CH4	2.2	2.2	2.2	5.1	5.2	5.9	6.0	5.7	5.1	5.1	4.5	5.0	2.3	2.0	4.5	4.2	4.7	4.7	4.7	
	CO2	17	18	17	38	39	41	41	40	37	37	42	36	18	19	43	40	40	40	41	
	N2	58	59	62	7	7	2	2	2	10	11	11	11	55	55	2	11	2	2	2	
	Total	100	100	100	100	100	100	100	100	100	100	100	100	100	100	100	100	100	100	100	
		26	27	27	24	24	24	24	23	24	24	26	24	26	27	24	25	23	23	24	
Heating Value, Btu/scf		97	89	82	213	211	224	227	229	209	205	186	205	103	99	211	188	222	221	217	
	Gasifier Temp C	784	808	789	776	782	781	774	778	782	798	800	784	771	792	762	782	760	753	750	
	Coal Feed Rate	374	337	445	445	440	457	567	571	534	502	487	586	480	456	413	469	504	531	499	
	Air Flow (lb/hr)	807	803	842	325	334	0	0	0	0	0	0	0	842	842	0	0	0	0	0	
	Steam Flow (lb/hr)	242	238	281	333	333	426	418	419	424	416	415	415	288	287	480	489	447	450	452	
	O2 flow (lb/hr)	0	0	0	131	131	211	216	228	217	213	209	211	0	0	204	201	210	209	207	
	Air/Coal Ratio (lb/lb)	2.34	2.59	2.06	0.79	0.82	0.00	0.00	0.00	0.00	0.00	0.00	0.00	2.16	2.28	0.00	0.00	0.00	0.00	0.00	
	Steam/Coal Ratio (lb/lb)	0.70	0.77	0.64	0.81	0.82	1.01	0.80	0.80	0.88	0.92	0.95	0.79	0.74	0.78	1.43	1.30	1.11	1.06	1.13	
	O2/Coal Ratio (lb/lb)	0.54	0.60	0.48	0.50	0.51	0.50	0.41	0.43	0.45	0.47	0.48	0.40	0.50	0.53	0.61	0.54	0.52	0.49	0.52	
	O2/maf Coal Ratio (lb/lb)	0.92	1.02	0.81	0.86	0.88	0.86	0.71	0.74	0.83	0.87	0.88	0.74	0.91	0.95	1.10	0.97	0.94	0.89	0.94	
	Recirc Rate (lb/hr)	1261	2827	2570	3256	4125	5193	7652	4583	4704	4003	4495	4965	2014	4168	2536	1226	1158	947	977	
	Fraction Carbon in Coal	0.40	0.40	0.40	0.40	0.40	0.40	0.40	0.40	0.37	0.37	0.37	0.38	0.38	0.38	0.38	0.38	0.38	0.38	0.38	
	Fraction Sorbent in Coal	8%	8%	8%	8%	8%	8%	8%	8%	10%	10%	10%	10%	19%	19%	19%	20%	20%	20%	20%	
	Coal Heating Value BTU/lb	6623	6623	6623	6623	6623	6623	6623	6623	6263	6263	6263	6263	6352	6352	6352	6352	6352	6352	6352	
	Wt WAtEr (lb/hr)	326	302	400	481	523	487	587	523	520	448	511	520	326	312	559	491	460	451	566	
	Wt LASH lb/hr	0	0	58	0	35	31	72	54	73	73	85	92	0	17	0	45	97	91	90	
	Fraction C in LASH	16.1%	6.0%	13.0%	13.0%	12.1%	12.9%	19.0%	8.6%	15.6%	9.6%	10.1%	15.5%	0.0%	35.7%	7.5%	12.0%	24.8%	10.4%	6.8%	
	WT filter ash lb/hr	33	28	36	52	41	39	49	55	53	48	40	46	154	55	73	47	64	56	57	
	Fraction C in filter ash	0.43	0.46	0.51	0.53	0.52	0.55	0.62	0.62	0.47	0.37	0.35	0.40	0.41	0.54	0.46	0.40	0.49	0.43	0.50	
	Wt dipleg lb/hr	0	0	0	0	0	0	0	0	0	6	0	8	8	31	0	27	0	0	0	
	Fraction C in dipleg	0.0%	0.0%	0.0%	0.0%	0.0%	0.0%	0.0%	0.0%	6.5%	0.0%	10.2%	8.5%	33.6%	0.0%	8.4%	0.0%	0.0%	0.0%	8.1%	
	TRDU Throughput lb/hr-ft^A	5151	4641	6129	6129	6060	6294	7809	7864	7195	6763	6561	7895	5820	5529	5008	5617	6036	6359	5976	
	TRDU Throughput MMBtu	34	31	41	41	40	42	52	45	42	42	41	49	37	35	32	36	38	40	38	
	TRDU Riser Vel ft/s	48	48	51	45	45	43	42	42	42	42	42	43	52	53	50	51	49	49	49	
	% Moisture As run	35	35	35	35	35	35	35	35	36	36	36	36	34	34	34	34	34	34	34	
	Ultimate As run																				
	C	40	40	40	40	40	40	40	40	37	37	37	37	38	38	38	38	38	38	38	
	H	6.8	6.8	6.8	6.8	6.8	6.8	6.8	6.8	6.6	6.6	6.6	6.6	6.2	6.2	6.2	6.2	6.2	6.2	6.2	
	N	1.5	1.5	1.5	1.5	1.5	1.5	1.5	1.5	1.4	1.4	1.4	1.4	1.5	1.5	1.5	1.5	1.5	1.5	1.5	
	S	1.1	1.1	1.1	1.1	1.1	1.1	1.1	1.1	1.1	1.1	1.1	1.1	3.1	3.1	3.1	3.1	3.1	3.1	3.1	
	O	44	44	44	44	44	44	44	44	44	44	44	44	40	40	40	40	40	40	40	
	Ash	6.3	6.3	6.3	6.3	6.3	6.3	6.3	6.3	9.4	9.4	9.4	9.4	10.9	10.9	10.9	10.9	10.9	10.9	10.9	
	Carbon Conversion																				
	Solid Accountability	90	90	85	83	85	85	79	82	80	85	86	83	57	67	74	82	64	79	77	
	Gas Make	79	88	69	80	81	85	74	79	86	88	98	76	91	94	101	90	93	86	93	
	Cold Gas Efficiency	41	40	31	44	43	48	43	48	50	50	51	45	53	50	56	48	58	52	56	
	Cold Gas Eff cor	41	41	31	45	44	49	44	48	50	50	49	44	51	49	56	47	57	52	54	
	Carbon Conv (calc)	74	81	79	86	85	88	84	83	94	89	90	88	111	100	57	88	86	87	94	
	PG HHV (GC - Btu/scf)	50	45	42	81	79	114	122	128	121	116	112	118	60	58	111	100	118	116	117	
	Hot Gas Eff (calc)	50	50	50	60	59	61	59	60	64	60	59	59	69	62	66	51	51	52	57	
	Cold Gas Eff. (calc)	36	35	34	46	44	48	47	49	51	47	46	47	51	45	52	38	42	42	46	
	H2S concetration (ppm)	1191	1401	1530	2158	2528	3524	3687	3561	3304	3493	2994	3039	1689	1596	967	2967	3218	3584	3709	
	ulfur Retention (%)	46	31	38	34	21	16	26	25	31	25	35	43	75	75	89	75	72	71	68	
	1-Lig, 2-Sub, 3-Bit, 4-Pet	1	1	1	1	1	1	1	1	1	1	1	1	1	1	1	1	1	1	1	
	TC416 thermocouple (402	606	663	688	682	708	719	711	690	665	674	690	666	621	698	612	604	576	552	539	

		Oxygen-blown lignite tests P068																			
		Center-1	Center-2	Center-3	Center-4	Center-5	Center-6	Center-7	Center-8	Falkirk-1	Falkirk-2	Falkirk-3	Falkirk-4	Freedom-1	Freedom-2	Freedom-3	HS Free-1	HS Free-2	HS Free-3	HS Free-4	
		06/12/2001	06/14/2001	06/14/2001	06/14/2001	06/15/2001	06/16/2001	06/16/2001	06/16/2001	06/16/2001	06/17/2001	06/17/2001	06/18/2001	06/18/2001	06/19/2001	06/20/2001	06/20/2001	06/21/2001	06/21/2001	06/21/2001	06/21/2001
		13:00-20:00	06:00-10:00	13:00-17:00	20:00-24:00	0:00-8:00	11:00-23:00	0:00-7:00	8:00-12:00	15:00-24:00	01:00-17:00	18:00-23:00	0:00-07:00	11:00-17:00	0:00-10:00	12:30-14:00	17:00-24:00	0:00-5:00	06:00-13:00	15:00-18:00	
SUMMARY OF THE DRY PRODUCT GAS																					
TOTAL PRODUCT GAS, scf/hr ----- >		20535	20008	20135	18977	18997	20408	20872	20123	20872	20872	20123	24134	22434	21524	20890	19894	19876	20128	20002	
	H2	7.1%	7.3%	8.1%	7.7%	7.7%	8.1%	8.2%	8.2%	8.7%	8.2%	8.3%	7.7%	8.5%	8.7%	8.2%	8.2%	8.5%	8.5%	8.5%	
	CO	4.4%	4.6%	4.7%	4.5%	4.5%	4.8%	4.9%	4.9%	5.0%	5.2%	5.2%	4.7%	4.6%	5.4%	5.4%	5.1%	5.4%	5.1%	5.4%	
	CH4	1.2%	1.1%	1.2%	1.1%	1.1%	1.1%	1.1%	1.1%	1.3%	1.4%	1.4%	1.3%	1.1%	1.1%	1.2%	1.1%	1.1%	1.1%	1.1%	
	CO2	13.3%	13.1%	13.0%	12.6%	12.4%	12.7%	12.7%	12.7%	12.9%	13.0%	13.4%	12.8%	12.6%	13.0%	13.2%	13.3%	13.3%	13.1%	13.0%	
	N2	74.4%	73.3%	72.7%	73.4%	73.8%	72.9%	72.8%	72.6%	71.3%	70.9%	71.4%	71.5%	72.1%	70.4%	69.9%	70.5%	70.1%	70.3%	70.2%	
	(ppm)	H2S	1088	1086	1078	1019	994	993	1017	901	800	800	800	446	875	935	946	906	852	941	917
SUMMARY OF THE ELEMENTAL AND MASS CLOSURE																					
	#MOLES H	4%	5%	11%	-11%	-13%	10%	3%	-10%	0%	-1%	-12%	4%	7%	8%	-7%	-1%	5%	-27%	4%	
	#MOLES C	3%	-18%	2%	-12%	0%	-1%	-7%	-25%	-13%	-15%	-33%	1%	8%	6%	7%	5%	2%	-2%	10%	
	#MOLES O	0%	0%	3%	-8%	-8%	4%	-1%	-7%	-2%	-4%	-10%	-4%	0%	2%	-6%	-3%	1%	-18%	0%	
	#MOLES S	0%	-1%	0%	0%	0%	0%	0%	-1%	0%	-1%	-1%	0%	1%	1%	1%	1%	0%	0%	1%	
	#MOLES N	1%	1%	2%	2%	2%	3%	3%	3%	1%	1%	1%	1%	0%	1%	1%	1%	1%	1%	1%	
	#MOLES Ca	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	
	#MOLES Mg	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	
	MASS, #	1%	1%	2%	1%	2%	2%	3%	3%	1%	1%	1%	1%	1%	1%	1%	1%	1%	1%	1%	
SUMMARY OF CARBON UTILIZATION																					
High Kinetics																					
% CARBON LOST BY GASIFICATION (Total)		13%	15%	13%	14%	14%	13%	12%	17%	19%	19%	27%	5%	5%	7%	8%	10%	10%	10%	11%	
% CARBON LOST BY COMBUSTION (Total)		73%	64%	66%	66%	72%	66%	67%	58%	60%	60%	50%	75%	81%	79%	81%	80%	74%	76%	81%	
% Carbon Removed In Filter Ash (Total)		12%	14%	19%	16%	11%	19%	19%	23%	17%	17%	20%	18%	12%	12%	10%	8%	15%	12%	7%	
% Carbon Removed In LASH (Total)		2%	7%	2%	4%	3%	2%	2%	2%	4%	4%	3%	1%	2%	1%	1%	1%	1%	3%	2%	
Char Carbon Accounted For (Comb + S.Gasif + Filter + LASH)		100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	
% Coal Carbon From Volatiles (Basis: Recycle Rate, Ultimate Analysis)		-8%	-34%	-4%	-24%	-11%	-10%	-12%	-44%	-25%	-25%	-67%	12%	15%	8%	6%	-0%	-2%	-6%	7%	
Single-Pass CHAR Carbon Conversion (Gasification)		5.3%	7.7%	5.2%	5.8%	5.2%	5.0%	4.6%	8.8%	4.7%	4.7%	9.0%	4.0%	4.3%	7.4%	8.2%	12.4%	9.4%	9.0%	8.0%	
Single-Pass CHAR Carbon Conversion (Combustion)		24%	24%	21%	22%	21%	20%	21%	23%	13%	13%	15%	37%	39%	45%	44%	49%	42%	41%	37%	
Bed Recycle Rate from Comb Zone nrg-Balance, lb/hr		3580	3358	3720	3333	3474	3905	3805	3424	3727	3745	3357	3757	3358	3176	3224	2825	3254	3348	3586	
Low Kinetics																					
% CARBON LOST BY GASIFICATION (Total)		4%	5%	4%	4%	5%	4%	4%	5%	6%	6%	6%	9%	2%	2%	2%	3%	3%	3%	3%	
% CARBON LOST BY COMBUSTION (Total)		80%	72%	73%	73%	80%	73%	73%	67%	69%	69%	63%	78%	84%	83%	86%	86%	80%	82%	88%	
% Carbon Removed In Filter Ash (Total)		13%	16%	21%	18%	13%	21%	21%	26%	20%	20%	24%	19%	13%	13%	10%	9%	16%	13%	7%	
% Carbon Removed In LASH (Total)		2%	7%	2%	5%	3%	2%	2%	2%	5%	5%	4%	1%	2%	2%	2%	2%	2%	3%	2%	
Char Carbon Accounted For (Comb + S.Gasif + Filter + LASH)		100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	
% Coal Carbon From Volatiles (Basis: Recycle Rate, Ultimate Analysis)		2%	-19%	5%	-12%	0%	1%	-2%	-26%	-8%	-8%	-34%	15%	19%	13%	12%	7%	5%	1%	15%	
Single-Pass CHAR Carbon Conversion (Gasification)		1.5%	2.0%	1.5%	1.7%	1.5%	1.4%	1.3%	2.3%	1.4%	1.4%	2.3%	1.2%	1.3%	2.0%	2.2%	3.1%	2.4%	2.3%	2.1%	
Single-Pass CHAR Carbon Conversion (Combustion)		24%	24%	21%	22%	21%	20%	21%	23%	13%	13%	15%	37%	39%	45%	44%	49%	42%	41%	37%	
Bed Recycle Rate from Comb Zone nrg-Balance, lb/hr		3580	3358	3720	3333	3474	3905	3805	3424	3727	3745	3357	3757	3358	3176	3224	2825	3254	3348	3586	
Summary of the Spreadsheets Results																					
H2O (Inlet)		9.7%	9.9%	10.0%	9.6%	9.3%	8.8%	8.6%	8.9%	8.9%	8.9%	8.9%	9.2%	9.1%	9.6%	10.1%	10.4%	10.8%	10.8%	10.7%	10.8%
H2O After Combustion (No WGS)		9.1%	9.3%	9.4%	9.0%	8.8%	8.3%	8.1%	8.3%	8.4%	8.4%	8.6%	8.6%	9.1%	9.5%	9.7%	10.1%	10.1%	10.0%	10.1%	
H2O After Combustion (WGS)		5.2%	5.4%	5.5%	5.2%	5.1%	4.7%	4.6%	4.7%	4.7%	4.8%	4.8%	5.0%	5.4%	5.5%	5.6%	5.9%	5.9%	5.9%	6.0%	
H2O After Steam-Gasification (WGS)		0.9%	0.7%	0.7%	0.8%	0.8%	0.6%	0.8%	0.5%	0.3%	0.3%	0.1%	2.0%	2.3%	2.0%	1.8%	1.8%	1.5%	1.5%	1.4%	
Summary of the Spreadsheets Results																					
Carbon Conversion (Coal IN - Filter/LASH) / (Coal IN)		85%	72%	78%	75%	85%	78%	76%	65%	74%	74%	62%	83%	88%	88%	90%	90%	84%	85%	92%	
Carbon Conversion (Coal IN - [1 - Balanced Product]) / (Coal IN)		83%	91%	77%	89%	86%	81%	86%	92%	87%	89%	96%	82%	80%	82%	83%	86%	82%	88%	83%	
% Sulfur On Sorbent, CHAR, & ASH (S IN - H2S) / (S IN)		84%	83%	85%	83%	84%	85%	84%	85%	88%	88%	87%	79%	59%	58%	58%	58%	62%	55%	59%	
Heat Loss from Mass & Energy Balances (H IN - H OUT) / (L)		-6%	-6%	-2%	-4%	-4%	-2%	-2%	-3%	-2%	-2%	-4%	-0%	-1%	-2%	-3%	-3%	-2%	-3%	-2%	
Heat Loss from TRDU (Btu/hr)		-1.5E+05	-1.3E+04	-4.8E+04	-9.0E+04	-8.3E+04	-4.4E+04	-4.6E+04	-6.8E+04	-5.9E+04	-5.6E+04	-8.1E+04	6.2E+03	-1.9E+04	-4.8E+04	-7.4E+04	-7.2E+04	-5.0E+04	-7.8E+04	-3.7E+04	
HHV of Product Gas, 60 F w H2S w/o tar, Btu/scf		49	50	54	51	51	53	54	54	58	60	58	55	53	57	59	56	58	57	57	
HHV of Product Gas, 60 F, w H2S w/o tar, Btu/scf (Dry Corrected)		117	122	133	138	139	133	136	135	141	143	139	132	130	127	128	126	131	128	132	
LHV of Product Gas, 60 F, w H2S w/o tar, Btu/scf (Wet)		41	42	45	44	43	45	46	47	50	51	50	47	44	48	49	46	48	47	48	
LHV of Product Gas, 60 F, w H2S w/o tar, Btu/scf (Wet Corrected)		80	85	89	93	93	92	96	96	101	103	101	94	89	88	88	86	88	89	90	
Gasifier Cold Gas Efficiency, % of coal HHV		41%	45%	41%	47%	46%	44%	47%	50%	49%	51%	52%	46%	44%	46%	47%	46%	46%	48%	46%	
Bed Recycle Rate from Comb Zone nrg-Balance, lb/hr		3580	3358	3720	3333	3474	3905	3805	3424	3727	3745	3357	3757	3358	3176	3224	2825	3254	3348	3586	
Comparison of Spreadsheets Results																					
delta T (°C)		178	145	101	94	74	62	63	88	117	124	110	118	150	94	150	178	184	201	211	
Average Mixing Zone T (°C)		784	808	789	776	782	781	774	778	782	798	800	784	771	792	762	782	760	753	750	
Average Low T (°C)		606	663	688	682	708	719	711	690	665	674	690	666	621	698	612	604	576	552	539	
Air In (scfh)		10,542	10,500	11,000	4,250	4,364	0	0	0	0	0	0	0	0	10,999	11,000	0	0	0	0	
Oxygen In (scfh)		0	0	0	1,549	1,549	2,504	2,563	2,702	2,570	2,526	2,474	2,502	0	0	2,412	2,379	2,488	2,478	2,449	
Nitrogen In (scfh)		4,082	3,991	4,676	4,063	4,065	4,123	4,123	4,146	4,088	4,140	4,133	4,198	4,324	4,340	5,380	3,998	4,030	4,062	4,002	
Product Gas (scfh)		19,452	19,044	20,612	15,627	15,638	12,503	13,075	13,768	13,110	12,611	12,341	12,920	21,392	21,419	13,495	11,623	12,941	12,868	12,537	
Coal Feed Rate (lb/hr)		374	337	445	445	440	457	567	571	534	502	487	586	480	456	413	469	504	531	499	
Air Flow (lb/hr)		807	803	842	325	334	0	0	0	0	0	0	0	842	842	0	0	0	0	0	
Steam Flow (lb/hr)		242	238	261	333	333	426	418	419	424	416	415	415	288	287	480	489	447	450	452	
Air/Coal Ratio (lb/lb)		2.34	2.59	2.06	0.79	0.82	0.00	0.00	0.00	0.00	0.00	0.00	0.00	2.16	2.28	0.00	0.00	0.00	0.00		

Gas Compositions from TRDU Test		P069											
Test		Wyodak	Wyodak	Wyodak	Wyodak	Wyodak	Wyodak	Wyodak	Wyodak	Wyodak	Wyodak	Wyodak	
Product Gas Comp, vol%		10/09/2001	10/10/2001	10/10/2001	10/10/2001	10/11/2001	10/12/2001	11/12/2001	10/13/2001	10/13/2001	10/13/2001	10/14/2001	10/14/2001
		04:38-07:00	12:00 - 10:01	14:00-18:25	18:25-01:00	02:00-8:00	12:00-20:00	22:00-03:00	04:00-15:00	16:00-03:00	03:00-18:00	22:00-08:00	
H2		8.6	8.7	15.2	14.8	17.4	10.0	16.1	19.3	19.4	20.0	19.1	
CO		4.0	4.4	6.5	5.9	6.8	5.7	7.1	7.8	8.7	10.4	11.0	
CH4		1.5	1.8	2.9	2.8	3.2	1.7	2.7	3.4	3.6	3.8	3.9	
CO2		14	14	20	20	24	14	19	24	24	24	24	
N2		81	81	62	64	54	70	61	49	48	44	45	
Total		109	110	106	107	105	101	106	104	103	103	104	
Ave Mol Wt		31	30	29	29	28	28	28	27	27	27	27	
Flow, scfh													
air in		9999	10961	3199	3084	0	12906	3200	0	0	0	0	
oxygen in		0	0	1695	1614	2219	0	1750	2400	2513	2798	2835	
nitrogen in		3904	4823	4908	5011	4991	5193	5118	5110	5091	5097	5080	
product gas		18578	21083	15866	15488	12900	25142	16356	14076	14543	15932	15718	
Product Gas, scfh													
H2		1600	1834	2412	2289	2250	2504	2628	2718	2814	3188	3004	
CO		745	932	1033	912	873	1421	1153	1101	1270	1657	1735	
CH4		279	371	457	440	415	427	440	474	516	605	611	
CO2		2638	2907	3153	3025	3071	3402	3162	3360	3477	3868	3813	
N2		15078	17119	9788	9835	6993	17524	9908	6961	6960	7069	7133	
Total		20339	23164	16842	16501	13603	25278	17292	14614	15037	16388	16296	
Heating Value, Btu/hr													
H2 (325)		519859	596122	783780	743966	731172	813847	854233	883375	914573	1036098	976206	
CO (321)		239138	299130	331553	292830	280339	455988	370144	353339	407543	531874	557021	
CH4 (1014)		282571	376256	463338	446017	421195	433398	446136	481002	523504	613892	619990	
Total		1041568	1271507	1578672	1482813	1432706	1703232	1670514	1717715	1845620	2181864	2153217	
Heating Value, Btu/scf													
H2		28	28	49	48	57	32	52	63	63	65	62	
CO		13	14	21	19	22	18	23	25	28	33	35	
CH4		15	18	29	29	33	17	27	34	36	39	39	
Total		56	60	100	96	111	68	102	122	127	137	137	
VALUES ADJUSTED FOR PURGE NITROGEN													
Product Gas Comp, vol%													
H2		9.7%	10.0%	20.2%	19.9%	26.1%	12.5%	21.6%	28.6%	28.3%	28.2%	26.8%	
CO		4.5%	5.1%	8.7%	7.9%	10.1%	7.1%	9.5%	11.6%	12.8%	14.7%	15.5%	
CH4		1.7%	2.0%	3.8%	3.8%	4.8%	2.1%	3.6%	5.0%	5.2%	5.4%	5.5%	
CO2		16.1%	15.9%	26.4%	26.3%	35.7%	16.9%	26.0%	35.4%	35.0%	34.3%	34.0%	
N2		68.0%	67.0%	40.9%	42.0%	23.2%	61.4%	39.4%	19.5%	18.8%	17.5%	18.3%	
Total		100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	
Ave Mol Wt		28	28	27	27	26	27	26	26	26	25	26	
Heating Value, Btu/scf													
		63	69	132	129	166	85	137	181	186	193	192	
Product Gas, scfh													
H2		1600	1834	2412	2289	2250	2504	2628	2718	2814	3188	3004	
CO		745	932	1033	912	873	1421	1153	1101	1270	1657	1735	
CH4		279	371	457	440	415	427	440	474	516	605	611	
CO2		2638	2907	3153	3025	3071	3402	3162	3360	3477	3868	3813	
N2		11174	12296	4880	4824	2002	12331	4790	1851	1869	1972	2053	
Total		16435	18341	11934	11490	8612	20085	12174	9504	9946	11291	11216	
FLUE GAS FLOWS ADJUSTED TO 450,000 BTU/HR HEAT LOSS													
Gas created due to combustion of													
Gas created due to combustion of		50	50	50	50	50	50	50	50	50	50	50	
CO2													
		779	779	779	779	779	779	779	779	779	779	779	
N2													
		3299	3299	3299	3299	3299	3299	3299	3299	3299	3299	3299	
Product Gas, scfh													
H2		1600	1834	2412	2289	2250	2504	2628	2718	2814	3188	3004	
CO		745	932	1033	912	873	1421	1153	1101	1270	1657	1735	
CH4		279	371	457	440	415	427	440	474	516	605	611	
CO2		1859	2128	2373	2245	2292	2622	2382	2580	2698	3089	3034	
N2		7875	8998	1581	1525	111	9032	1492	120	126	140	142	
Total		12357	14263	7856	7412	5941	16007	8096	6994	7423	8679	8526	

	P069											
	Wyodak	Wyodak	Wyodak	Wyodak	Wyodak	Wyodak	Wyodak	Wyodak	Wyodak	Wyodak	Wyodak	
	10/09/2001	10/10/2001	10/10/2001	10/10/2001	10/11/2001	10/12/2001	11/12/2001	10/13/2001	10/13/2001	10/13/2001	10/14/2001	10/14/2001
	04:38-07:00	12:00 - 10:01	04:00-18:25	18:25-01:00	02:00-8:00	12:00-20:00	22:00-03:00	04:00-15:00	16:00-03:00	03:00-18:00	22:00-08:00	
Product Gas, vol %												
H2	13	13	31	31	38	16	33	39	38	37	35	
CO	6.0	6.5	13.1	12.3	14.7	8.9	14.2	15.7	17.1	19.1	20.4	
CH4	2.3	2.6	5.8	5.9	7.0	2.7	5.4	6.8	7.0	7.0	7.2	
CO2	15	15	30	30	39	16	29	37	36	36	36	
N2	64	63	20	21	2	56	18	2	2	2	2	
Total	100	100	100	100	100	100	100	100	100	100	100	
	27	27	24	24	23	26	24	23	23	23	24	
Heating Value, Btu/scf	84	89	201	200	241	106	206	246	249	251	253	
Gasifier Temp C	804	823	835	832	828	804	823	827	840	865	892	
Coal Feed Rate	281	385	390	400	349	476	376	381	400	456	406	
Air Flow (lb/hr)	765	839	245	236	0	988	245	0	0	0	0	
Steam Flow (lb/hr)	297	288	325	341	392	269	342	405	361	365	364	
O2 flow (lb/hr)	0	0	143	136	187	0	148	203	212	236	239	
Air/Coal Ratio (lb/lb)	2.84	2.27	0.65	0.61	0.00	2.16	0.68	0.00	0.00	0.00	0.00	
Steam/Coal Ratio (lb/lb)	1.10	0.78	0.87	0.89	1.17	0.59	0.95	1.11	0.94	0.83	0.93	
O2/Coal Ratio (lb/lb)	0.66	0.53	0.53	0.50	0.56	0.50	0.57	0.55	0.55	0.54	0.61	
O2/mat Coal Ratio (lb/lb)	0.91	0.73	0.74	0.69	0.78	0.69	0.79	0.77	0.77	0.75	0.85	
Recirc Rate (lb/hr)	2700	2750	3010	3675	3140	1530	1340	1790	2780	3045	3665	
Fraction Carbon in Coal	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	
Fraction Sulfur in Coal	4%	4%	4%	4%	4%	4%	4%	4%	4%	4%	4%	
Coal Heating Value BTU/lb	9010	9010	9010	9010	9010	9010	9010	9010	9010	9010	9010	
Wt WATER (lb/hr)	414	365	397	521	418	357	388	436	351	386	368	
Wt LASH lb/hr	0	0	0	0	0	0	0	0	0	11	0	
Fraction C in LASH	12.5%	5.9%	13.4%	21.0%	12.8%	41.2%	24.3%	18.7%	13.7%	9.3%	6.8%	
Wt filter ash lb/hr	87	79	63	65	39	75	48	34	38	35	31	
Fraction C in filter ash	0.48	0.61	0.67	0.64	0.66	0.70	0.64	0.57	0.56	0.52	0.33	
Wt dipleg lb/hr	20	8	3	0	0	16	0	0	0	0	0	
Fraction C in dipleg	1.8%	5.5%	9.4%	0.0%	0.0%	12.5%	0.0%	0.0%	0.0%	0.0%	0.0%	
TRDU Throughput lb/hr-ft ²	4038	5533	5605	5749	5016	6841	5404	5475	5749	6553	5835	
TRDU Throughput MMBtu/	36	50	50	52	45	62	49	49	52	59	53	
TRDU Riser Vel ft/s	46	51	42	42	40	55	43	42	41	43	44	
% Moisture As run	23	23	23	23	23	23	23	23	23	23	23	
Ultimate As run												
C	58	58	58	58	58	58	58	58	58	58	58	
H	6.2	6.2	6.2	6.2	6.2	6.2	6.2	6.2	6.2	6.2	6.2	
N	0.7	0.7	0.7	0.7	0.7	0.7	0.7	0.7	0.7	0.7	0.7	
S	0.4	0.4	0.4	0.4	0.4	0.4	0.4	0.4	0.4	0.4	0.4	
O	30	30	30	30	30	30	30	30	30	30	30	
Ash	4.9	4.9	4.9	4.9	4.9	4.9	4.9	4.9	4.9	4.9	4.9	
Carbon Conversion												
Solid Accountability	73	77	80	81	87	79	85	91	90	92	95	
Gas Make	74	62	68	62	71	63	72	74	75	76	86	
Cold Gas Efficiency	47	42	50	46	50	42	54	54	55	57	64	
Cold Gas Eff cor	43	38	47	43	47	41	51	52	53	55	61	
Carbon Conv (calc)	68	60	73	68	67	79	76	82	80	83	96	
PG HHV (GC - Btu/scf)	59	63	106	102	118	71	108	130	134	144	141	
Hot Gas Eff (calc)	48	44	55	52	50	59	59	62	60	62	71	
Cold Gas Eff. (calc)	34	33	45	41	40	46	48	51	51	53	60	
	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	
	0	0	0	0	0	0	0	0	0	0	0	
H2S concentration (ppm)	262	424	766	776	854	500	683	868	891	1009	1020	
sulfur Retention (%)	58	43	24	27	23	35	27	22	21	14	4	
1-Lig, 2-Sub, 3-Bit, 4-Pet	2	2	2	2	2	2	2	2	2	2	2	
TC416 thermocouple (402	658	679	668	661	646	639	638	639	683	718	764	

SUMMARY OF THE DRY PRODUCT GAS

TOTAL PRODUCT GAS, scf/hr ----->	20567	21242	20711	20299	21443	19993	19680	19002	20689	19603	16772
H2	9.0%	9.1%	8.3%	7.4%	8.7%	8.3%	7.8%	8.2%	8.7%	8.3%	7.2%
CO	5.4%	5.2%	5.3%	4.9%	5.3%	4.6%	4.0%	3.7%	5.9%	6.5%	5.7%
CH4	1.2%	1.1%	1.0%	1.0%	1.1%	1.0%	0.9%	1.0%	1.3%	1.1%	1.0%
CO2	12.9%	12.7%	12.5%	12.9%	12.8%	12.5%	12.0%	12.2%	13.8%	12.9%	13.0%
N2	69.8%	70.3%	71.2%	72.3%	70.2%	71.3%	72.8%	72.2%	74.0%	74.7%	75.8%
{ ppm}	H2S	815	978	840	797	694	866	773	886	229	627

SUMMARY OF THE ELEMENTAL AND MASS CLOSURE

#MOLES H	8%	11%	2%	-9%	6%	10%	11%	10%	18%	10%	24%
#MOLES C	13%	14%	2%	4%	13%	5%	2%	3%	15%	13%	2%
#MOLES O	3%	4%	0%	-8%	1%	3%	4%	5%	10%	5%	14%
#MOLES S	2%	2%	0%	1%	2%	1%	0%	1%	2%	2%	0%
#MOLES N	1%	1%	1%	1%	1%	1%	1%	1%	1%	1%	1%
#MOLES Ca	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%
#MOLES Mg	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%
MASS, #	1%	1%	1%	1%	1%	1%	1%	1%	1%	1%	0%

SUMMARY OF CARBON UTILIZATION

High Kinetics

% CARBON LOST BY GASIFICATION { Total }	6%	5%	8%	10%	7%	6%	5%	3%	13%	14%	23%
% CARBON LOST BY COMBUSTION { Total }	82%	82%	84%	77%	84%	76%	69%	75%	68%	73%	68%
% Carbon Removed in Filter Ash { Total }	10%	13%	7%	13%	9%	18%	16%	28%	7%	8%	5%
% Carbon Removed in LASH { Total }	3%	1%	0%	0%	0%	0%	4%	0%	5%	5%	4%
Char Carbon Accounted For { Comb + S.Gasif + Filter + LASH }	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%
% Coal Carbon From Volatiles (Basis: Recycle Rate, Ultimate Analy)	18%	21%	-4%	-9%	14%	6%	4%	3%	8%	4%	-31%
Single-Pass CHAR Carbon Conversion { Gasification }	4.6%	3.8%	10.2%	16.9%	6.8%	5.6%	5.5%	2.9%	5.4%	5.5%	11.5%
Single-Pass CHAR Carbon Conversion { Combustion }	40%	39%	51%	56%	45%	43%	47%	37%	23%	22%	25%
Bed Recycle Rate from Comb Zone nrg-Balance, lb/hr	4452	4538	3788	3523	4238	3937	3280	4202	3729	3659	3260

Low Kinetics

% CARBON LOST BY GASIFICATION { Total }	2%	1%	2%	3%	2%	2%	1%	4%	4%	7%
% CARBON LOST BY COMBUSTION { Total }	85%	84%	89%	83%	88%	79%	79%	70%	83%	81%
% Carbon Removed in Filter Ash { Total }	10%	14%	8%	14%	9%	19%	16%	28%	7%	8%
% Carbon Removed in LASH { Total }	3%	1%	1%	0%	1%	1%	4%	0%	6%	5%
Char Carbon Accounted For { Comb + S.Gasif + Filter + LASH }	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%
% Coal Carbon From Volatiles (Basis: Recycle Rate, Ultimate Analy)	21%	23%	3%	-0%	19%	10%	7%	5%	17%	14%
Single-Pass CHAR Carbon Conversion { Gasification }	1.3%	1.1%	2.5%	3.9%	1.8%	1.5%	1.6%	0.9%	1.5%	1.6%
Single-Pass CHAR Carbon Conversion { Combustion }	40%	39%	51%	56%	45%	43%	47%	37%	23%	22%
Bed Recycle Rate from Comb Zone nrg-Balance, lb/hr	4452	4538	3788	3523	4238	3937	3280	4202	3729	3659

{ Inlet }	10.6%	10.3%	10.3%	10.4%	10.1%	10.6%	10.7%	11.0%	10.4%	10.8%	11.0%
{ No WGS }	9.9%	9.7%	9.6%	9.6%	9.5%	10.0%	10.1%	10.4%	9.7%	10.1%	10.2%
{ WGS }	5.8%	5.7%	5.6%	5.5%	5.5%	6.0%	6.2%	6.3%	5.6%	5.9%	5.9%
{ WGS }	2.0%	2.1%	1.7%	1.5%	1.8%	2.2%	2.8%	2.6%	0.8%	0.7%	0.4%

SUMMARY OF THE SPREADSHEET RESULTS

Carbon Conversion (Coal IN - Filter/LASH) / (Coal IN)	90%	89%	92%	86%	92%	83%	82%	73%	89%	86%	88%
Carbon Conversion (Coal IN - [1 - Balanced Product]) / (Coal IN)	77%	75%	90%	82%	80%	77%	80%	70%	75%	75%	86%
% Sulfur On Sorbent, CHAR, & ASH (S IN - H2S) / (S IN)	66%	60%	58%	63%	70%	61%	61%	61%	93%	79%	73%
Heat Loss from Mass & Energy Balances (H IN - H Out) / (H IN)	-1%	-1%	-4%	-3%	-1%	-2%	-1%	-3%	-2%	-8%	
Heat Loss from TRDU (Btu/hr)	-3.5E+04	-1.7E+04	-9.4E+04	-6.1E+04	-2.9E+04	-4.2E+04	-2.7E+04	-6.7E+04	-8.0E+04	-6.0E+04	-1.3E+05
HHV of Product Gas, 60 F w H2S w/o tar, Btu/scf	60	59	55	50	58	53	49	50	58	57	50
HHV of Product Gas, 60 F, w H2S w/o tar, Btu/scf {Dry Corrected}	134	135	124	115	129	131	136	133	125	131	119
LHV of Product Gas, 60 F, w H2S w/o tar, Btu/scf {Wet}	49	49	46	41	48	44	41	40	46	45	41
LHV of Product Gas, 60 F, w H2S w/o tar, Btu/scf {Wet Corrected}	90	91	86	77	88	86	88	81	81	82	75
Gasifier Cold Gas Efficiency, % of coal HHV	44%	44%	50%	42%	45%	42%	43%	39%	41%	41%	43%
Bed Recycle Rate from Comb Zone nrg-Balance, lb/hr	4452	4538	3788	3523	4238	3937	3280	4202	3729	3659	3260

COMPARISON OF SPREADSHEET RESULTS

delta T (°C)	146	144	167	171	182	165	185	188	157	147	128
Average Mixing Zone T (°C)	804	823	835	832	828	804	823	827	840	865	892
Average Low T (°C)	658	679	668	661	646	639	638	639	683	718	764
Air in (scfh)	9,999	10,961	3,199	3,084	0	12,906	3,200	0	0	0	0
Oxygen in (scfh)	0	0	1,695	1,614	2,219	0	1,750	2,400	2,513	2,798	2,835
Nitrogen in (scfh)	3,904	4,823	4,908	5,011	4,991	5,193	5,118	5,110	5,091	5,097	5,080
Product Gas (scfh)	18,578	21,083	15,866	15,488	12,900	25,142	16,356	14,076	14,543	15,932	15,718
Coal Feed Rate (lb/hr)	281	381	390	400	349	476	376	381	400	456	406
Air Flow (lb/hr)	765	839	245	236	0	988	245	0	0	0	0
Steam Flow (lb/hr)	297	288	325	341	392	269	342	405	361	365	364
Air/Coal Ratio (lb/lb)	2.84	2.27	0.65	0.61	0.00	2.16	0.68	0.00	0.00	0.00	0.00
Steam/Coal Ratio (lb/lb)	1.10	0.78	0.87	0.89	1.17	0.59	0.95	1.11	0.94	0.83	0.83
O2/Coal Ratio (lb/lb)	0.68	0.53	0.53	0.50	0.56	0.50	0.57	0.55	0.55	0.54	0.61
Fraction Carbon in Coal	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58
Bed Recycle Rate, lb/hr (by heat balance around burner)	4,452	4,538	3,788	3,523	4,238	3,937	3,280	4,202	3,729	3,659	3,260
Bed Recycle Rate, lb/hr (Calculated)	2,700	2,750	3,010	3,675	3,140	1,530	1,340	1,790	2,780	3,045	3,665

	P070																			Illinois No.	Illinois No.	Illinois No.	Illinois No.
	SUFCo	SUFCo	SUFCo	SUFCo	SUFCo	SUFCo	SUFCo	SUFCo	SUFCo	SUFCo	SUFCo	SUFCo	SUFCo	SUFCo	SUFCo	SUFCo	SUFCo	SUFCo	SUFCo				
	04/15/2002	04/15/2002	04/15/2002	04/16/2002	04/16/2002	04/16/2002	04/17/2002	04/17/2002	04/17/2002	04/17/2002	04/17/2002	04/17/2002	04/17/2002	04/18/2002	04/18/2002	04/18/2002	04/19/2002	04/19/2002	04/20/2002				
	16:00-18:00	18:00-20:00	20:00-24:00	0:00-05:00	05:00-10:00	11:00-16:00	18:00-20:00	0:00-5:00	07:00-12:00	12:00-18:00	20:00-23:00	0:00-07:00	08:00-11:00	12:00-13:45	12:00-16:30	18:00-04:30	05:00-07:00	07:30-13:30	13:30-14:45	17:00-19:00			
Product Gas, vol %																							
H2	14	13	14	17	17	15	52	43	41	35	36	36	37	37	36	34	35	30	37	25			
CO	6.9	6.4	7.1	7.5	7.3	8.0	11.8	15.7	14.9	17.7	17.5	15.2	15.3	15.4	14.7	16.0	16.5	15.2	17.1	16.0			
CH4	2.7	2.8	3.0	3.1	2.9	2.7	5.6	7.7	7.5	8.5	8.6	8.8	9.4	9.3	9.0	8.2	8.5	7.2	8.0	6.7			
CO2	17	18	16	16	15	16	31	33	36	38	36	39	37	36	38	39	37	45	35	49			
N2	60	60	60	56	58	59	0	0	1	1	1	1	2	2	3	3	3	3	3	3			
Total	100	100	100	100	100	100	100	100	100	100	100	100	100	100	100	100	100	100	100	100			
	27	27	27	26	26	26	19	21	22	24	23	24	23	23	24	24	24	24	27	28			
Heating Value, Btu/scf	96	90	98	112	106	102	263	268	256	257	261	255	265	265	255	245	254	218	257	201			
Gasifier Temp C	900	903	898	880	885	918	856	886	894	908	906	893	885	897	867	896	900	944	972	972			
Coal Feed Rate	258	290	313	412	313	318	327	357	332	298	276	300	331	279	326	301	296	220	295	295			
Air Flow (lb/hr)	873	903	903	907	1014	765	941	317	309	227	247	0	0	0	0	0	0	57	57	56			
Steam Flow (lb/hr)	287	287	288	287	288	288	351	382	383	382	385	434	430	452	460	459	454	457	463	458			
O2 flow (lb/hr)	0	0	0	0	0	0	18	96	124	127	130	187	191	183	210	211	211	211	223	217			
Air/Coal Ratio (lb/lb)	3.52	3.24	3.01	2.29	3.13	3.32	2.44	1.29	1.00	1.08	0.86	0.00	0.00	0.00	0.00	0.00	0.00	0.31	0.23	0.23			
Steam/Coal Ratio (lb/lb)	1.16	1.03	0.96	0.73	0.96	0.94	1.12	1.11	1.20	1.34	1.45	1.51	1.35	1.69	1.47	1.59	1.60	2.50	1.89	1.87			
O2/Coal Ratio (lb/lb)	0.82	0.75	0.70	0.53	0.73	0.77	0.62	0.58	0.62	0.69	0.69	0.65	0.60	0.68	0.67	0.73	0.74	1.23	0.96	0.94			
O2/maf Coal Ratio (lb/lb)	0.98	0.91	0.84	0.64	0.88	0.93	0.75	0.70	0.75	0.84	0.83	0.78	0.72	0.83	0.81	0.88	0.90	1.52	1.19	1.16			
Recirc Rate (lb/hr)	1395	2080	1990	4120	5555	4440	1540	2835	2060	1610	1745	3800	5870	2040	6780	4995	6740	5490	4615	4235			
Fraction Carbon in Coal	0.70	0.70	0.70	0.70	0.70	0.70	0.70	0.70	0.70	0.70	0.70	0.70	0.70	0.70	0.70	0.70	0.70	0.63	0.63	0.63			
Fraction Sorbent in Coal	4%	4%	4%	4%	4%	4%	4%	4%	4%	4%	4%	4%	4%	4%	4%	4%	4%	17%	17%	17%			
Coal Heating Value BTU/lb	11600	11600	11600	11600	11600	11600	11600	11600	11600	11600	11600	11600	11600	11600	11600	11600	11600	11300	11300	11300			
Wt Water (lb/hr)	251	263	267	273	281	299	366	332	312	306	295	360	369	368	436	429	422	408	386	466			
Wt LASH lb/hr	0	0	0	0	0	0	0	0	0	0	0	0	0	5	13	0	0	0	0	0			
Fraction C in LASH	7.8%	7.8%	7.8%	12.7%	5.8%	13.0%	8.8%	11.2%	21.3%	32.6%	22.5%	15.7%	28.4%	27.8%	19.4%	4.1%	6.5%	0.8%	0.5%	0.7%			
WT filter ash lb/hr	50	95	180	105	87	91	340	101	89	89	104	61	80	154	100	88	62	96	85	68			
Fraction C in filter ash	0.33	0.60	0.68	0.63	0.66	0.52	0.61	0.65	0.56	0.60	0.59	0.63	0.65	0.50	0.62	0.50	0.55	0.39	0.35	0.30			
Wt dipleg lb/hr	0	0	86	0	115	0	44	73	45	35	28	82	0	0	0	0	0	0	0	0			
Fraction C in dipleg	0.0%	0.0%	1.2%	0.0%	0.0%	0.0%	0.0%	0.0%	0.0%	0.0%	0.0%	23.5%	0.0%	0.0%	0.0%	0.0%	0.0%	0.0%	0.0%	0.0%			
TRDU Throughput lb/hr-ft*	3708	4168	4498	5921	4498	4570	4699	5131	4771	4283	3969	4311	4757	4010	4685	4326	4254	2734	3665	3665			
TRDU Throughput MMBtu/	43	48	52	69	52	53	55	60	55	50	46	50	55	47	54	50	49	31	41	41			
TRDU Riser Vel ft/s	55	57	57	57	53	56	53	51	51	51	49	46	45	47	54	56	57	60	61	61			
% Moisture As run	10	10	10	10	10	10	10	10	10	10	10	10	10	10	10	10	10	9	9	9			
Ultimate As run																							
C	70	70	70	70	70	70	70	70	70	70	70	70	70	70	70	70	70	63	63	63			
H	5.2	5.2	5.2	5.2	5.2	5.2	5.2	5.2	5.2	5.2	5.2	5.2	5.2	5.2	5.2	5.2	5.2	5.6	5.6	5.6			
N	1.2	1.2	1.2	1.2	1.2	1.2	1.2	1.2	1.2	1.2	1.2	1.2	1.2	1.2	1.2	1.2	1.2	1.0	1.0	1.0			
S	0.3	0.3	0.3	0.3	0.3	0.3	0.3	0.3	0.3	0.3	0.3	0.3	0.3	0.3	0.3	0.3	0.3	3.2	3.2	3.2			
O	16	16	16	16	16	16	16	16	16	16	16	16	16	16	16	16	16	16	16	16			
Ash	7.6	7.6	7.6	7.6	7.6	7.6	7.6	7.6	7.6	7.6	7.6	7.6	7.6	7.6	7.6	7.6	7.6	10.7	10.7	10.7			
Carbon Conversion																							
Solid Accountability	90	71	41	76	73	78	6	72	78	73	67	81	68	59	71	78	83	68	81	87			
Gas Make	60	56	53	42	54	57	58	53	66	73	70	77	61	73	51	56	56	103	57	69			
Cold Gas Efficiency	35	31	31	28	37	36	57	43	54	55	51	54	46	56	35	36	38	57	38	31			
Cold Gas Eff cor	35	30	31	28	37	36	51	41	48	49	48	52	43	53	34	35	37	52	36	29			
Carbon Conv (calc)	82	82	74	77	77	84	74	78	95	105	100	111	107	123	102	104	113	142	148	122			
PG HHV (GC - Btu/scf)	58	56	60	70	68	66	90	99	122	117	108	134	142	143	114	109	115	107	115	70			
Hot Gas Eff (calc)	59	54	53	60	61	62	74	66	74	76	76	82	81	96	74	71	80	72	70	54			
Cold Gas Eff. (calc)	43	40	40	47	48	48	60	55	63	64	63	70	70	82	63	59	67	58	56	41			
	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97			
	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0			
H2S concetration (ppm)	228	223	315	213	307	273	339	323	404	701	784	600	716	786	599	796	792	2070	2000	2368			
ulfur Retention (%)	59	64	51	74	50	55	42	60	51	11	-5	29	34	11	46	24	24	69	79	70			
1-Lig, 2-Sub, 3-Bit, 4-Pet	3	3	3	3	3	3	3	3	3	3	3	3	3	3	3	3	3	3	3	3			
TC416 thermocouple (402	773	810	810	815	833	860	777	803	793	794	795	795	792	747	802	828	843	884	905	908			

Gas Compositions from TRUDU Test

P071

Pet Coke Pet Coke Pet Coke Prater CreelPrater CreelPrater CreelPrater CreelPrater Creel

Test 06/10/2002 06/13/2002 06/13/2002 06/14/2002 06/14/2002 06/14/2002 06/15/2002 06/15/2002 06/15/2002 06/15/2002

Product Gas Comp, vol% 16:00-20:00 10:00-14:00 16:00-20:00 5:00-7:39 09:00-19:00 20:00-04:00 04:00-12:00 13:00-20:00 20:00-01:00

	H2	6.7	19.0	18.8	13.9	14.8	14.2	14.7	14.2	14.1
	CO	5.6	9.2	9.9	7.8	7.2	7.4	7.6	7.8	7.7
	CH4	0.6	1.6	1.6	3.3	3.4	3.2	3.4	3.0	2.9
	CO2	12	22	20	17	17	18	18	16	15
	N2	74	49	50	58	59	62	61	61	62
	Total	99	101	101	100	102	105	104	102	102
	Ave Mol Wt	28	27	26	27	27	28	28	27	27

Flow, scfh	air in	13400	0	750	750	750	750	750	750	750
	oxygen in	0	2606	2562	2493	2492	2497	2514	2791	2649
	nitrogen in	5548	5823	5403	5445	5480	5487	5482	5323	5388
	product gas	20983	11601	12152	11970	12422	12163	12704	13606	13180

Product Gas, scfh	H2	1412	2205	2288	1660	1840	1728	1861	1935	1854
	CO	1179	1066	1198	935	892	902	966	1065	1008
	CH4	130	180	191	393	420	387	429	408	384
	CO2	2436	2541	2446	2078	2096	2188	2246	2152	2036
	N2	15576	5709	6104	6922	7379	7559	7726	8357	8181
	Total	20734	11700	12227	11988	12626	12765	13228	13917	13463

Heating Value, Btu/hr	H2 (325)	458955	716714	743669	539576	597905	561721	604892	628781	602673
	CO (321)	378541	342217	384617	300088	286301	289702	309938	341966	323647
	CH4 (1014)	131917	182327	193457	398112	425744	392201	435423	413881	388898
	Total	969414	1241257	1321743	1237776	1309950	1243624	1350253	1384629	1315218

Heating Value, Btu/scf	H2	22	62	61	45	48	46	48	46	46
	CO	18	29	32	25	23	24	24	25	25
	CH4	6	16	16	33	34	32	34	30	30
	Total	46	107	109	103	105	102	106	102	100

VALUES ADJUSTED FOR PURGE NITROGEN

Product Gas Comp, vol%	H2	9.3%	37.5%	33.5%	25.4%	25.7%	23.7%	24.0%	22.5%	23.0%
	CO	7.8%	18.1%	17.6%	14.3%	12.5%	12.4%	12.5%	12.4%	12.5%
	CH4	0.9%	3.1%	2.8%	6.0%	5.9%	5.3%	5.5%	4.7%	4.7%
	CO2	16.0%	43.2%	35.8%	31.8%	29.3%	30.1%	29.0%	25.0%	25.2%
	N2	66.0%	-1.9%	10.3%	22.6%	26.6%	28.5%	29.0%	35.3%	34.6%
	Total	100%	100%	100%	100%	100%	100%	100%	100%	100%
	Ave Mol Wt	28	25	25	26	25	26	26	26	25

Heating Value, Btu/scf	64	211	194	189	183	171	174	161	163
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Product Gas, scfh	H2	1412	2205	2288	1660	1840	1728	1861	1935	1854
	CO	1179	1066	1198	935	892	902	966	1065	1008
	CH4	130	180	191	393	420	387	429	408	384
	CO2	2436	2541	2446	2078	2096	2188	2246	2152	2036
	N2	10028	-114	701	1477	1899	2072	2244	3034	2793
	Total	15186	5877	6824	6543	7146	7278	7746	8594	8075

FLUE GAS FLOWS ADJUSTED TO 450,000 BTU/HR HEAT LOSS

Gas created due to combustion of										
Gas created due to combustion of		29	29	29	35	35	35	35	35	35
	CO2	451	451	451	547	547	547	547	547	547
	N2	1907	1907	1907	2313	2313	2313	2313	2313	2313

Product Gas, scfh	H2	1412	2205	2288	1660	1840	1728	1861	1935	1854
	CO	1179	1066	1198	935	892	902	966	1065	1008
	CH4	130	180	191	393	420	387	429	408	384
	CO2	1985	2090	1996	1531	1549	1641	1699	1606	1490
	N2	8121	130	721	717	717	717	718	732	725
	Total	12828	5671	6393	5236	5418	5376	5674	5746	5461

	P071									
	Pet Coke	Pet Coke	Pet Coke	Prater Creel	Prater Creel	Prater Creel	Prater Creel	Prater Creel	Prater Creel	Prater Creel
	06/10/2002	06/13/2002	06/13/2002	06/14/2002	06/14/2002	06/14/2002	06/14/2002	06/15/2002	06/15/2002	06/15/2002
	16:00-20:00	10:00-14:00	16:00-20:00	05:00-7:39	09:00-19:00	02:00-04:00	04:00-12:00	13:00-20:00	20:00-01:00	
Product Gas, vol %										
H2	11	39	36	32	34	32	33	34	34	34
CO	9.2	18.8	18.7	17.9	16.5	16.8	17.0	18.5	18.5	18.5
CH4	1.0	3.2	3.0	7.5	7.8	7.2	7.6	7.1	7.0	7.0
CO2	16	37	31	29	29	31	30	28	27	27
N2	63	2	11	14	13	13	13	13	13	13
Total	100	100	100	100	100	100	100	100	100	100
	27	23	23	24	23	24	23	23	23	23
Heating Value, Btu/scf	76	219	207	236	242	231	238	241	241	241
Gasifier Temp C	1020	943	965	974	955	960	955	980	973	973
Coal Feed Rate	289	208	241	247	312	286	303	329	308	308
Air Flow (lb/hr)	1025	0	57	57	57	57	57	57	57	57
Steam Flow (lb/hr)	261	456	456	456	456	456	456	473	467	467
O2 flow (lb/hr)	0	220	216	210	210	211	212	236	224	224
Air/Coal Ratio (lb/lb)	4.73	0.00	0.32	0.25	0.20	0.21	0.20	0.19	0.20	0.20
Steam/Coal Ratio (lb/lb)	1.20	2.92	2.52	1.96	1.56	1.70	1.60	1.53	1.61	1.61
O2/Coal Ratio (lb/lb)	1.10	1.41	1.27	0.96	0.76	0.83	0.79	0.80	0.82	0.82
O2/maf Coal Ratio (lb/lb)	1.12	1.44	1.29	1.12	0.88	0.97	0.92	0.93	0.95	0.95
Recirc Rate (lb/hr)	2261	3900	2729	2152	3800	6105	5322	5352	5022	5022
Fraction Carbon in Coal	0.90	0.90	0.90	0.71	0.71	0.71	0.71	0.71	0.71	0.71
Fraction Sorbent in Coal	25%	25%	25%	6%	6%	6%	6%	6%	6%	6%
Coal Heating Value BTU/lb	15584	15584	15584	12847	12847	12847	12847	12847	12847	12847
Wt Water (lb/hr)	214	370	388	424	440	430	412	442	428	428
Wt LASH lb/hr	0	0	0	0	0	18	0	0	0	0
Fraction C in LASH	11.6%	28.2%	32.8%	9.2%	9.5%	4.9%	7.2%	10.9%	10.8%	10.8%
WT filter ash lb/hr	141	69	77	88	129	90	101	91	97	97
Fraction C in filter ash	0.30	0.44	0.48	0.59	0.62	0.64	0.65	0.62	0.64	0.64
Wt dipleg lb/hr	0	0	0	10	0	4	0	0	0	0
Fraction C in dipleg	15.5%	0.0%	30.3%	19.0%	9.0%	8.2%	0.0%	0.0%	0.0%	0.0%
TRDU Throughput lb/hr-ft ²	3245	2339	2706	3477	4388	4023	4265	4634	4330	4330
TRDU Throughput MMBtu/hr	51	36	42	45	56	52	55	60	56	56
TRDU Riser Vel ft/s	54	62	64	65	64	65	65	67	67	67
% Moisture As run	1	1	1	7	7	7	7	7	7	7
Ultimate As run										
C	90	90	90	71	71	71	71	71	71	71
H	4.0	4.0	4.0	5.2	5.2	5.2	5.2	5.2	5.2	5.2
N	1.7	1.7	1.7	1.5	1.5	1.5	1.5	1.5	1.5	1.5
S	5.4	5.4	5.4	0.7	0.7	0.7	0.7	0.7	0.7	0.7
O	-2	-2	-2	15	15	15	15	15	15	15
Ash	1.0	1.0	1.0	6.6	6.6	6.6	6.6	6.6	6.6	6.6
Carbon Conversion										
Solid Accountability	78	78	77	68	61	69	68	74	70	70
Gas Make	58	81	71	65	51	57	56	52	52	52
Cold Gas Efficiency	28	51	47	42	35	38	38	36	36	36
Cold Gas Eff cor	29	51	47	41	35	36	37	35	35	35
Carbon Conv. (calc)	129	124	107	150	147	146	134	126	126	126
PG HHV (GC - Btu/scf)	46	107	109	104	106	103	104	102	98	98
Hot Gas Eff. (calc)	44	65	58	73	72	75	69	65	65	65
Cold Gas Eff. (calc)	31	54	49	60	60	61	57	54	53	53
	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97
	0	0	0	0	0	0	0	0	0	0
H2S concentration (ppm)	758	4620	4798	1776	1613	1484	1416	1320	1259	1259
sulfur Retention (%)	89	47	50	-10	18	19	24	30	31	31
1-Lig, 2-Sub, 3-Bit, 4-Pet	4	4	4	3	3	3	3	3	3	3
TC416 thermocouple (402	904	839	880	880	879	897	888	910	903	903

	P072a				P072B																									
	Illinois No. 6	Illinois No. 6	Illinois No. 6	Illinois No. 6	Illinois No. 6	Illinois No. 6	Illinois No. 6	Illinois No. 6	Illinois No. 6	Illinois No. 6	Illinois No. 6	Illinois No. 6	Illinois No. 6	Pittsburgh N	Pittsburgh N	Calumet	Calumet	Calumet	Calumet	Calumet	Calumet	Calumet	Calumet	Calumet	Calumet	Calumet	Calumet	Calumet		
	10/08/2002	10/08/2002	10/08/2002	10/09/2002	10/09/2002	10/09/2002	10/09/2002	10/21/2002	10/21/2002	10/22/2002	10/22/2002	10/22/2002	10/22/2002	10/23/2002	10/23/2002	10/23/2002	10/23/2002	10/23/2002	10/24/2002	10/24/2002	10/24/2002	10/24/2002	10/24/2002	10/24/2002	10/24/2002	10/24/2002	10/24/2002	10/24/2002	10/24/2002	10/25/2002
	01:00-05:00	06:00-09:00	10:00-14:00	15:00-19:00	20:00-24:00	01:00-05:00	06:00-09:00	10:00-14:00	15:00-19:00	20:00-24:00	01:00-05:00	06:00-09:00	10:00-14:00	15:00-19:00	20:00-24:00	01:00-05:00	06:00-09:00	10:00-14:00	15:00-19:00	20:00-24:00	01:00-05:00	06:00-09:00	10:00-14:00	15:00-19:00	20:00-24:00	01:00-05:00	06:00-09:00	10:00-14:00	15:00-19:00	20:00-24:00
Product Gas, vol %																														
H2	5	6	12	29	31	27	28	28	31	33	26	28	29	30	30	33	32	32	32	32	32	32	32	32	32	32	32	32	33	
CO	3.6	3.8	6.2	15.2	15.3	14.2	15.7	15.1	16.6	13.8	12.1	13.3	14.1	14.5	14.8	18.3	16.5	17.7	16.8	17.5	18.2									
CH4	1.3	1.4	3.0	7.8	6.1	5.6	5.6	5.0	5.6	4.6	4.5	6.5	6.6	7.8	7.7	6.8	6.3	7.1	6.5	6.3	6.0									
CO2	10	10	17	33	33	39	34	35	31	32	40	36	35	34	34	30	33	31	32	32	31									
N2	80	79	63	15	15	14	16	17	16	17	18	17	15	13	14	12	13	13	12	12	12									
Total	100	100	100	100	100	100	100	100	100	100	100	100	100	100	100	100	100	100	100	100	100									
	28	28	27	25	25	27	25	26	24	24	27	26	25	25	25	23	24	24	24	24	24									
Heating Value, Btu/scf	42	47	88	223	210	189	199	191	210	197	171	199	207	224	223	235	220	232	224	225	227									
Gasifier Temp C	964	974	941	934	990	969	1011	1028	1016	1030	1035	957	953	940	949	987	989	982	987	1006	1015									
Coal Feed Rate	269	253	306	334	281	289	283	279	294	382	255	244	281	311	305	351	322	293	305	300	327									
Air Flow (lb/hr)	899	899	421	57	57	54	57	57	57	63	88	68	68	54	54	54	54	54	54	54	54									
Steam Flow (lb/hr)	262	262	416	416	425	442	436	436	438	439	438	437	435	441	440	434	435	438	437	437	435									
O2 flow (lb/hr)	0	0	118	194	209	188	232	223	229	211	190	184	186	188	194	228	210	214	216	229	232									
Air/Coal Ratio (lb/lb)	4.03	4.28	1.65	0.21	0.25	0.22	0.24	0.25	0.23	0.18	0.31	0.30	0.28	0.19	0.19	0.16	0.18	0.20	0.19	0.19	0.18									
Steam/Coal Ratio (lb/lb)	1.18	1.25	1.64	1.50	1.82	1.76	1.86	1.88	1.79	1.28	1.98	1.93	1.67	1.53	1.55	1.33	1.45	1.61	1.54	1.57	1.43									
O2/Coal Ratio (lb/lb)	0.93	0.99	0.85	0.75	0.95	0.84	1.04	1.02	0.99	0.66	0.93	0.88	0.77	0.69	0.73	0.74	0.74	0.83	0.80	0.87	0.80									
O2/maf Coal Ratio (lb/lb)	1.16	1.23	1.05	0.93	1.18	1.03	1.29	1.27	1.23	0.73	1.03	1.09	0.96	0.86	0.91	0.92	1.03	1.00	1.07	1.00										
Recirc Rate (lb/hr)	1910	2162	2005	4804	4510	3829	8464	8272	7314	6842	5954	6623	6323	4595	6375	6613	5768	6289	6578	6483	6794									
Fraction Carbon in Coal	0.63	0.63	0.63	0.63	0.63	0.63	0.63	0.63	0.63	0.76	0.76	0.67	0.67	0.67	0.67	0.67	0.67	0.67	0.67	0.67	0.67									
Fraction Sorbent in Coal	17%	17%	17%	17%	17%	17%	17%	17%	17%	10%	13%	7%	7%	7%	7%	7%	7%	7%	7%	7%	7%									
Coal Heating Value BTU/lb	11300	11300	11300	11300	11300	11300	11300	11300	11300	13327	13327	12214	12214	12214	12214	12214	12214	12214	12214	12214	12214									
Wt Water (lb/hr)	236	232	429	371	378	371	390	394	401	375	357	349	351	347	347	371	354	348	357	363	369									
Wt LASH lb/hr	0	0	0	0	0	0	0	0	28	0	0	0	0	0	0	0	62	4	0	0	0									
Fraction C in LASH	1.0%	3.9%	3.6%	4.7%	1.0%	0.4%	1.2%	0.3%	0.9%	0.2%	0.5%	2.5%	3.0%	3.7%	3.2%	2.6%	2.1%	5.1%	4.8%	4.3%	4.3%									
Wt filter ash lb/hr	62	67	120	119	73	183	59	64	61	95	56	88	109	91	107	109	76	98	93	97	86									
Fraction C in filter ash	0.34	0.29	0.30	0.46	0.30	0.24	0.44	0.24	0.36	0.19	0.32	0.51	0.57	0.61	0.56	0.56	0.52	0.53	0.53	0.53	0.53									
Wt dipleg lb/hr	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0	0									
Fraction C in dipleg																														
TRDU Throughput lb/hr-ft ²	3336	3147	3807	4152	3496	3593	3518	3464	3658	5151	3316	3390	3905	4323	4239	4887	4476	4079	4246	4170	4546									
TRDU Throughput MMBtu/t	38	36	43	47	40	41	40	39	41	69	44	41	48	53	52	60	55	50	52	51	56									
TRDU Riser Vel ft/s	67	68	65	56	60	59	62	62	62	62	62	59	58	58	59	60	60	59	59	60	60									
% Moisture As run	9	9	9	9	9	9	9	9	9	2	2	3	3	3	3	3	3	3	3	3	3									
Ultimate As run																														
C	63	63	63	63	63	63	63	63	63	76	76	67	67	67	67	67	67	67	67	67	67									
H	5.6	5.6	5.6	5.6	5.6	5.6	5.6	5.6	5.6	5.4	5.4	4.3	4.3	4.3	4.3	4.3	4.3	4.3	4.3	4.3	4.3									
N	1.0	1.0	1.0	1.0	1.0	1.0	1.0	1.0	1.0	1.4	1.4	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9	1.9									
S	3.2	3.2	3.2	3.2	3.2	3.2	3.2	3.2	3.2	3.2	1.6	1.6	0.7	0.7	0.7	0.7	0.7	0.7	0.7	0.7	0.7									
O	16	16	16	16	16	16	16	16	16	8	8	10	10	10	10	10	10	10	10	10	10									
Ash	10.7	10.7	10.7	10.7	10.7	10.7	10.7	10.7	10.7	7.3	7.3	16.1	16.1	16.1	16.1	16.1	16.1	16.1	16.1	16.1	16.1									
Carbon Conversion																														
Solid Accountability	85	85	77	69	85	71	83	89	86	93	90	70	64	72	68	72	80	71	74	73	77									
Gas Make	49	50	46	58	67	68	63	62	61	34	57	68	63	56	55	55	55	61	60	63	58									
Cold Gas Efficiency	18	21	22	35	38	33	34	31	35	20	26	35	34	32	31	35	31	38	35	36	34									
Cold Gas Eff cor	18	21	22	35	39	33	34	31	35	20	27	35	34	32	32	35	32	38	35	38	36									
Carbon Conv (calc)	38	61	81	103	108	99	103	107	94	93	99	93	126	97	93	105	107	130	122	125	127									
PG HHV (GC - Btu/scf)	23	26	46	96	85	76	73	66	76	76	70	88	99	104	99	120	104	115	109	112	113									
Hot Gas Eff (calc)	21	36	45	62	63	50	65	66	64	64	56	59	82	67	65	77	74	85	85	88	91									
Cold Gas Eff. (calc)	11	21	31	50	50	38	49	49	49	48	41	46	65	54	52	64	60	69	69	72	75									
	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97	0.97									
H2S concentration (ppm)	0	157	1112	2461	3420	3241	3009	3447	3291	3462	2809	1664	1665	1708	1594	1593	1636	1802	1487	1498	1475									
Sulfur Retention (%)	100	96	84	73	54	58	58	51	55	35	23	2	15	22	26	31	26	12	29	26	32									
1-Lig, 2-Sub, 3-Bit, 4-Pet	3	3	3	3	3	3	3	3	3	3	3	3	3	3	3	3	3	3	3	3	3									

Gas Compositions from TRDU Test		P075										
Test		Loy Yang	Loy Yang	Loy Yang	Loy Yang	Loy Yang	Lochiel w/d	Lochiel w/d	Lochiel w/d	Lochiel w/d	Lochiel w/d	Lochiel w/d
Product Gas Comp, vol%		12/01/2003	12/01/2003	12/01/2003	12/02/2003	12/03/2003	12/03/2003	12/04/2003	12/05/2003	12/05/2003	12/05/2003	12/05/2003
		13:20-16:00	16:00-19:20	19:20-22:00	00:35-00:50	18:45-20:45	23:00-23:20	20:11-24:11	02:30-04:40	04:40-06:40	06:40-08:40	11:30-13:30
H2		6.6	7.4	7.0	12.5	9.3	7.5	4.3	6.1	6.3	5.6	13.8
CO		4.4	5.0	5.3	8.3	6.0	3.5	3.3	4.4	4.6	4.7	4.9
CH4		1.4	1.7	1.9	3.5	3.1	0.8	1.1	1.5	1.5	1.5	3.5
CO2		12	13	12	18	22	19	14	14	14	13	25
N2		74	74	72	58	58	78	79	74	75	75	56
Total		99	100	99	100	98	109	101	100	101	99	103
Ave Mol Wt		28	28	28	27	28	32	29	28	29	28	29
Flow, scfh												
air in		10010	10010	10057	0	0	0	10036	11657	13127	13005	0
oxygen in		0	0	0	2128	2281	1911	0	0	0	0	1994
nitrogen in		4813	4714	4733	4959	6485	6420	6674	6260	6212	6336	5951
product gas		14211	14532	15367	9786	10045	11210	16041	18290	21089	23089	11247
Product Gas, scfh												
H2		938	1075	1076	1223	934	841	690	1116	1329	1293	1552
CO		625	727	814	812	603	392	529	805	970	1085	551
CH4		199	247	292	343	311	90	176	274	316	346	394
CO2		1762	1846	1890	1732	2210	2164	2198	2542	2931	2955	2778
N2		10474	10681	11080	5715	5786	8721	12608	13535	15796	17201	6298
Total		13998	14576	15152	9825	9844	12208	16201	18272	21342	22881	11573
Heating Value, Btu/hr												
H2 (325)		304826	349495	349599	397556	303610	273244	224173	362599	431797	420220	504428
CO (321)		200716	233239	261439	260728	193467	125944	169922	258328	311400	348344	176904
CH4 (1014)		201739	250503	296061	347305	315755	90936	178921	278191	320764	351184	399156
Total		707281	833236	907099	1005590	812831	490124	573017	899118	1063961	1119747	1080488
Heating Value, Btu/scf												
H2		21	24	23	41	30	24	14	20	20	18	45
CO		14	16	17	27	19	11	11	14	15	15	16
CH4		14	17	19	35	31	8	11	15	15	15	35
Total		50	57	59	103	81	44	36	49	50	48	96
VALUES ADJUSTED FOR PURGE NITROGEN												
Product Gas Comp, vol%												
H2		10.2%	10.9%	10.3%	25.1%	20.5%	14.5%	7.2%	9.3%	8.8%	7.8%	27.6%
CO		6.8%	7.4%	7.8%	16.7%	13.2%	6.8%	5.6%	6.7%	6.4%	6.6%	9.8%
CH4		2.2%	2.5%	2.8%	7.0%	6.8%	1.5%	1.9%	2.3%	2.1%	2.1%	7.0%
CO2		19.2%	18.7%	18.1%	35.6%	48.5%	37.4%	23.1%	21.2%	19.4%	17.9%	49.4%
N2		61.6%	60.5%	60.9%	15.5%	11.0%	39.8%	62.3%	60.6%	63.3%	65.7%	6.2%
Total		100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%
Ave Mol Wt		28	28	28	26	30	30	30	29	29	29	28
Heating Value, Btu/scf												
		77	84	87	207	178	85	60	75	70	68	192
Product Gas, scfh												
H2		938	1075	1076	1223	934	841	690	1116	1329	1293	1552
CO		625	727	814	812	603	392	529	805	970	1085	551
CH4		199	247	292	343	311	90	176	274	316	346	394
CO2		1762	1846	1890	1732	2210	2164	2198	2542	2931	2955	2778
N2		5661	5967	6347	756	500	2301	5934	7275	9584	10865	347
Total		9185	9862	10419	4866	4558	5788	9527	12012	15130	16545	5622
FLUE GAS FLOWS ADJUSTED TO 450,000 BTU/HR HEAT LOSS												
Gas created due to combustion of												
Gas created due to combustion of		47	47	47	47	47	61	61	61	61	61	61
CO2												
		735	735	735	735	735	950	950	950	950	950	950
N2												
		3110	3110	3110	0	0	0	4022	4022	4022	4022	0
Product Gas, scfh												
H2		938	1075	1076	1223	934	841	690	1116	1329	1293	1552
CO		625	727	814	812	603	392	529	805	970	1085	551
CH4		199	247	292	343	311	90	176	274	316	346	394
CO2		1027	1111	1155	997	1475	1213	1247	1592	1981	2005	1828
N2		2550	2857	3237	756	300	2301	1912	3252	5562	6843	347
Total		5340	6017	6574	4131	3623	4837	4555	7039	10157	11573	4672

		P075											
		Loy Yang	Loy Yang	Loy Yang	Loy Yang	Loy Yang	Lochiel w/d	Lochiel w/d	Lochiel w/d	Lochiel w/d	Lochiel w/d	Lochiel w/d	Lochiel w/d
		12/01/2003	12/01/2003	12/01/2003	12/02/2003	12/03/2003	12/03/2003	12/04/2003	12/05/2003	12/05/2003	12/05/2003	12/05/2003	12/05/2003
		13:20-16:00	16:00-19:20	19:20-22:00	00:35-00:50	18:45-20:45	23:00-23:20	20:11-24:11	02:30-04:40	04:40-06:40	06:40-08:40	11:30-13:30	
Product Gas, vol %													
	H2	18	18	16	30	26	17	15	16	13	11	33	
	CO	11.7	12.1	12.4	19.7	16.6	8.1	11.6	11.4	9.6	9.4	11.8	
	CH4	3.7	4.1	4.4	8.3	8.6	1.9	3.9	3.9	3.1	3.0	8.4	
	CO2	19	19	18	24	41	25	27	23	20	17	39	
	N2	48	48	49	18	8	48	42	46	55	59	7	
	Total	100	100	100	100	100	100	100	100	100	100	100	
		26	26	26	23	27	27	28	27	27	28	25	
Heating Value, Btu/scf		132	138	138	243	224	101	126	128	105	97	231	
	Gasifier Temp C	893	882	865	877	795	661	753	781	785	785	741	
	Coal Feed Rate	275	267	308	342	443	367	419	406	479	550	482	
	Air Flow (lb/hr)	766	766	770	0	0	0	768	892	1004	995	0	
	Steam Flow (lb/hr)	126	125	139	311	319	324	124	124	124	167	293	
	O2 flow (lb/hr)	0	0	0	180	193	161	0	0	0	0	168	
	Air/Coal Ratio (lb/lb)	2.79	2.87	2.50	0.00	0.00	0.00	2.26	2.71	2.59	2.23	0.00	
	Steam/Coal Ratio (lb/lb)	0.46	0.47	0.45	0.91	0.72	1.09	0.37	0.38	0.32	0.37	0.61	
	O2/Coal Ratio (lb/lb)	0.65	0.66	0.58	0.53	0.43	0.52	0.63	0.60	0.60	0.52	0.35	
	O2/maf Coal Ratio (lb/lb)	0.76	0.78	0.68	0.62	0.51	0.78	0.75	0.90	0.86	0.74	0.50	
	Recirc Rate (lb/hr)	10136	11879	11930	12295	2214	2657	8720	7729	8307	8015	11224	
	Fraction Carbon in Coal	0.56	0.56	0.56	0.56	0.56	0.46	0.46	0.46	0.46	0.46	0.46	
	Fraction Sorbent in Coal	0%	0%	0%	0%	0%	19%	19%	19%	19%	19%	0%	
	Coal Heating Value BTU/lb	9556	9556	9556	9556	9556	7389	7389	7389	7389	7389	7389	
	Wt WATER (lb/hr)	186	174	186	200	347	270	206	221	218	227	305	
	Wt LASH lb/hr	0	0	0	0	0	0	0	0	0	0	0	
	Fraction C in LASH	1.5%	2.2%	4.4%	0.0%	7.0%	8.6%	4.1%	3.2%	1.4%	4.9%	8.9%	
	WT filter ash lb/hr	61	48	69	70	61	123	118	128	138	108	94	
	Fraction C in filter ash	0.60	0.71	0.81	0.84	0.69	0.36	0.36	0.40	0.36	0.38	0.60	
	Wt dipleg lb/hr						0						
	Fraction C in dipleg						4.2%						
	TRDU Throughput lb/hr-ft ²	4117	3997	4611	5120	6632	4450	5081	4923	5808	6669	7216	
	TRDU Throughput MMBtu/hr	39	38	44	49	63	33	38	36	43	49	53	
	TRDU Riser Vel ft/s	36	36	38	34	36	35	34	37	42	45	30	
	% Moisture As run	14	14	14	14	14	18	18	18	18	18	18	
	Ultimate As run												
	C	56	56	56	56	56	46	46	46	46	46	46	
	H	5.5	5.5	5.5	5.5	5.5	5.5	5.5	5.5	5.5	5.5	5.5	
	N	0.7	0.7	0.7	0.7	0.7	0.6	0.6	0.6	0.6	0.6	0.6	
	S	0.3	0.3	0.3	0.3	0.3	3.0	3.0	3.0	3.0	3.0	3.0	
	O	36	36	36	36	36	33	33	33	33	33	33	
	Ash	1.0	1.0	1.0	1.0	1.0	12.3	12.3	12.3	12.3	12.3	12.3	
	Carbon Conversion												
	Solid Accountability	76	77	68	70	83	67	73	66	72	80	74	
	Gas Make	53	59	55	47	40	57	55	71	70	64	53	
	Cold Gas Efficiency	27	33	30	31	19	24	23	37	38	34	31	
	Cold Gas Eff cor	27	33	31	31	19	22	23	37	37	34	30	
	Carbon Conv. (calc)	72	64	73	81	106	84	69	127	141	105	101	
	PG HHV (GC - Btu/scf)	50	79	79	104	82	45	38	52	53	51	104	
	Hot Gas Eff (calc)	48	59	67	62	63	42	31	66	74	55	56	
	Cold Gas Eff. (calc)	37	49	56	53	52	30	23	52	59	44	49	
	H2S concentration (ppm)	313	276	310	587	515	3750	2190	1976	1722	1970	9013	
	ulfur Retention (%)	60	63	62	58	71	60	70	69	73	71	40	
	1-Lig, 2-Sub, 3-Bit, 4-Pet	1	1	1	1	1	1	1	1	1	1	1	
	TC416 thermocouple (402	823	822	810	800	652	423	697	735	748	756	701	

SUMMARY OF THE DRY PRODUCT GAS												
TOTAL PRODUCT GAS, scf/hr ----- >		15627	15638	12503	13075	13768	13110	12611	12341	12920	21392	21419
H2		13.5%	12.7%	18.5%	19.5%	20.8%	19.8%	18.8%	18.3%	19.5%	9.3%	8.6%
CO		4.8%	4.7%	6.1%	6.6%	7.4%	7.0%	6.9%	6.8%	6.9%	4.7%	4.7%
CH4		1.9%	1.9%	2.8%	3.1%	3.1%	2.8%	2.7%	2.6%	2.8%	1.3%	1.1%
CO2		20.8%	20.9%	28.4%	28.9%	28.8%	28.9%	28.8%	33.3%	28.6%	15.6%	15.3%
N2		57.1%	57.0%	42.8%	41.5%	39.0%	40.4%	42.6%	43.8%	42.6%	73.0%	72.1%
{ ppm}	H2S	2158	2528	3524	3687	3561	3304	3493	2994	3039	1689	1596

SUMMARY OF THE ELEMENTAL AND MASS CLOSURE												
#MOLES H		-5%	-12%	7%	-2%	5%	4%	13%	5%	7%	12%	15%
#MOLES C		1%	1%	-2%	5%	2%	-7%	-2%	-8%	7%	-30%	-25%
#MOLES O		-7%	-13%	2%	-6%	0%	-1%	7%	-4%	2%	5%	5%
#MOLES S		0%	0%	-1%	1%	0%	-1%	-1%	-1%	1%	-1%	-1%
#MOLES N		0%	0%	0%	0%	0%	0%	0%	-1%	0%	1%	1%
#MOLES Ca		0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%
#MOLES Mg		0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%
MASS, #		0%	0%	0%	0%	0%	0%	0%	0%	0%	1%	1%

SUMMARY OF CARBON UTILIZATION High Kinetics												
% CARBON LOST BY GASIFICATION { Total }		7%	7%	8%	9%	4%	8%	6%	7%	8%	3%	20%
% CARBON LOST BY COMBUSTION { Total }		75%	76%	76%	67%	74%	71%	78%	78%	71%	61%	56%
% Carbon Removed in Filter Ash { Total }		18%	14%	14%	17%	20%	15%	11%	9%	11%	36%	15%
% Carbon Removed in LASH { Total }		0%	3%	3%	8%	3%	7%	5%	6%	9%	0%	9%
Char Carbon Accounted For { Comb + S.Gasif + Filter + LASH }		100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%
% Coal Carbon From Volatiles {Basis: Recycle Rate, Ultimate Analy}		7%	6%	8%	14%	18%	5%	8%	7%	16%	-20%	-39%
Single-Pass CHAR Carbon Conversion { Gasification }		1.5%	1.3%	1.0%	0.9%	1.1%	1.8%	2.5%	2.4%	1.9%	2.4%	3.1%
Single-Pass CHAR Carbon Conversion { Combustion }		14%	12%	9%	6%	17%	13%	23%	20%	14%	32%	8%
Bed Recycle Rate from Comb Zone nrg-Balance, lb/hr		6107	8031	10660	10566	8607	5780	5385	5907	5509	3387	3870

Low Kinetics												
% CARBON LOST BY GASIFICATION { Total }		2%	2%	3%	3%	1%	3%	2%	2%	3%	1%	7%
% CARBON LOST BY COMBUSTION { Total }		79%	80%	80%	71%	76%	75%	81%	82%	75%	62%	65%
% Carbon Removed in Filter Ash { Total }		19%	15%	15%	18%	20%	15%	12%	10%	12%	36%	18%
% Carbon Removed in LASH { Total }		0%	3%	3%	8%	3%	7%	5%	6%	10%	0%	10%
Char Carbon Accounted For { Comb + S.Gasif + Filter + LASH }		100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%
% Coal Carbon From Volatiles {Basis: Recycle Rate, Ultimate Analy}		11%	11%	12%	19%	20%	10%	12%	12%	21%	-17%	-20%
Single-Pass CHAR Carbon Conversion { Gasification }		0.5%	0.4%	0.3%	0.3%	0.4%	0.6%	0.8%	0.7%	0.6%	0.8%	1.0%
Single-Pass CHAR Carbon Conversion { Combustion }		14%	12%	9%	6%	17%	13%	23%	20%	14%	32%	8%
Bed Recycle Rate from Comb Zone nrg-Balance, lb/hr		6107	8031	10660	10566	8607	5780	5385	5907	5509	3387	3870

{H2O} { Inlet }		37.9%	37.7%	53.5%	52.7%	52.4%	53.4%	52.9%	53.8%	52.6%	25.9%	25.7%
{H2O} After Combustion { No WGS }		35.6%	35.4%	49.8%	49.1%	48.7%	49.6%	49.2%	49.9%	49.0%	24.7%	24.5%
{H2O} After Combustion { WGS }		27.7%	27.5%	40.3%	39.4%	38.7%	39.9%	39.4%	40.1%	39.4%	18.9%	18.7%
{H2O} After Steam-Gasification { WGS }		1.9%	1.0%	0.5%	0.2%	2.8%	1.8%	3.5%	2.4%	1.6%	6.2%	0.3%

SUMMARY OF THE SPREADSHEET RESULTS												
Carbon Conversion { Coal In - Filter/LASH } / { Coal In }		84%	85%	85%	79%	82%	80%	86%	87%	83%	59%	68%
Carbon Conversion { Coal In - [1 - Balanced Product] } / { Coal In }		82%	84%	86%	74%	79%	87%	87%	93%	76%	89%	94%
% Sulfur On Sorbent, CHAR, & ASH { S IN - H2S } / { S IN }		32%	19%	15%	25%	24%	30%	26%	39%	43%	75%	75%
Heat Loss from Mass & Energy Balances { H IN - H Out } / { Heat Loss from TRDU { Btu/hr }		-8%	-8%	-9%	-5%	-6%	-6%	-7%	-7%	-3%	0%	1%
HHV of Product Gas, 60 F w H2S w/o tar, Btu/scf		80	79	111	117	125	114	109	114	114	57	54
HHV of Product Gas, 60 F, w H2S w/o tar, Btu/scf { Dry Corrected }		226	224	228	230	232	230	229	209	230	131	125
LHV of Product Gas, 60 F, w H2S w/o tar, Btu/scf { Wet }		51	50	60	63	69	64	61	57	61	42	40
LHV of Product Gas, 60 F, w H2S w/o tar, Btu/scf { Wet Corrected }		87	86	83	86	93	88	84	79	83	72	70
Gasifier Cold Gas Efficiency, % of coal HHV		45%	45%	49%	43%	48%	51%	50%	46%	44%	49%	48%
Bed Recycle Rate from Comb Zone nrg-Balance, lb/hr		6107	8031	10660	10566	8607	5780	5385	5907	5509	3387	3870

COMPARISON OF SPREADSHEET RESULTS												
delta T (°C)		70	60	55	77	143	238	56	46	37	29	40
Average Mixing Zone T (°C)		893	882	865	877	795	661	753	781	785	785	741
Average Low T (°C)		823	822	810	800	652	423	697	735	748	756	701
Air in (scfh)		10,010	10,010	10,057	0	0	0	10,036	11,657	13,127	13,005	0
Oxygen in (scfh)		0	0	0	2,128	2,281	1,911	0	0	0	0	1,994
Nitrogen in (scfh)		4,813	4,714	4,733	4,959	6,485	6,420	6,674	6,260	6,212	6,336	5,951
Product Gas (scfh)		14,211	14,532	15,367	9,786	10,045	11,210	16,041	18,290	21,089	23,089	11,247
Coal Feed Rate (lb/hr)		275	267	308	342	443	367	419	406	479	550	482
Air Flow (lb/hr)		768	766	770	0	0	0	768	892	1004	995	0
Steam Flow (lb/hr)		128	125	139	311	319	324	124	124	124	167	293
Air/Coal Ratio (lb/lb)		2.79	2.87	2.50	0.00	0.00	0.00	2.28	2.71	2.59	2.23	0.00
Steam/Coal Ratio (lb/lb)		0.46	0.47	0.45	0.91	0.72	1.09	0.37	0.38	0.32	0.37	0.61
O2/Coal Ratio (lb/lb)		0.56	0.66	0.58	0.53	0.43	0.54	0.52	0.63	0.60	0.52	0.35
Fraction Carbon in Coal		0.56	0.56	0.56	0.56	0.56	0.46	0.46	0.46	0.46	0.46	0.46
Bed Recycle Rate, lb/hr (by heat balance around burner)		6,107	8,031	10,660	10,566	8,607	5,780	5,385	5,907	5,509	3,387	3,870
Bed Recycle Rate, lb/hr (Calculated)		10,136	11,879	11,930	12,295	2,214	2,657	8,720	7,729	8,307	8,015	11,224

10-Dec-04	RUN NUMBER	Wyodak w/d	Wyodak w/d	Wyodak w/d	Wyodak w/d	Wyodak w/d	Wyodak w/d	Wyodak w/d	Wyodak w/d	Wyodak w/d	Wyodak w/d	Wyodak w/d	Wyodak w/d	Wyodak w/d	Wyodak w/d	Wyodak w/d	Wyodak w/d	Wyodak w/d	Wyodak w/d	Wyodak w/d	Wyodak w/d	Wyodak w/d	Wyodak w/d	
Balancing Was Performed On The Following :		06/01/2004	06/02/2004	06/02/2004	06/02/2004	06/02/2004	06/03/2004	06/03/2004	06/03/2004	06/04/2004	06/04/2004	06/04/2004	06/04/2004	06/05/2004	06/05/2004	06/06/2004	06/06/2004	06/06/2004	06/06/2004	06/07/2004	06/07/2004	06/07/2004	06/08/2004	06/08/2004
O - H2OOUT, N - Dry Product Gas, C - Conversion, H - N/A		06/13/2001	06/13/2001	06/13/2001	06/13/2001	06/13/2001	06/13/2001	06/13/2001	06/13/2001	06/13/2001	06/13/2001	06/13/2001	06/13/2001	06/13/2001	06/13/2001	06/13/2001	06/13/2001	06/13/2001	06/13/2001	06/13/2001	06/13/2001	06/13/2001	06/13/2001	06/13/2001
Balanced by Difference :		Type of Coal																						
		Subbit	Subbit	Subbit	Subbit	Subbit	Subbit	Subbit	Subbit	Subbit	Subbit	Subbit	Subbit	Subbit	Subbit	Subbit	Subbit	Subbit	Subbit	Subbit	Subbit	Subbit	Subbit	Subbit
O2/Coal Ratio (wt%)		0.77	0.69	0.65	0.58	0.59	0.63	0.62	0.64	0.70	0.85	0.74	0.68	0.85	0.69	0.69	0.63	0.57	0.70	0.64	0.69	0.68	0.64	0.75
Fraction Carbon in Coal		0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58	0.58
Bed Recycle Rate, lb/hr (by heat balance)		5.001	10.820	13.580	11.579	9.301	7.547	8.607	8.533	7.727	8.018	4.517	4.885	5.260	5.519	6.491	8.118	9.100	10.114	8.987	9.347	10.124	10.194	10.113
Bed Recycle Rate, lb/hr (Calc from dp's)		3.588	7.277	8.841	7.714	6.594	5.647	6.278	7.033	7.553	6.565	4.844	5.594	5.703	5.356	5.537	4.831	9.754	11.393	10.741	11.778	11.454	12.540	11.813

Fraction Sorbent in Coal	4%	4%	4%	4%	4%	4%	4%	4%	4%	4%	4%	4%
Coal Heating Value BTU/lb	6555	6555	6555	6555	6555	6568	6568	6568	6568	6568	6568	6568
WT filter ash lb/hr	187	184	179	177	190	137	200	146	129	110	141	
Fraction C in filter ash	0.19	0.16	0.13	0.11	0.13	0.16	0.12	0.14	0.28	0.28	0.20	
Wt dipleg lb/hr	0	0	0	0	0	0	0	0	0	0	0	
Fraction C in dipleg												
TRDU Throughput lb/hr-ft ²	6022	6683	6438	5978	6481	6338	5490	7257	5978	6223	6625	
TRDU Throughput MMBtu/h	39	44	42	39	42	42	36	48	39	41	44	
TRDU Riser Vel ft/s	50	51	51	50	46	48	49	56	38	39	41	
% Moisture As run	5	5	5	5	5	9	9	9	9	9	9	
Ultimate As run												
C	44	44	44	44	44	41	41	41	41	41	41	
H	3.7	3.7	3.7	3.7	3.7	3.9	3.9	3.9	3.9	3.9	3.9	
N	1.1	1.1	1.1	1.1	1.1	1.0	1.0	1.0	1.0	1.0	1.0	
S	0.5	0.5	0.5	0.5	0.5	0.6	0.6	0.6	0.6	0.6	0.6	
O	12	12	12	12	12	18	18	18	18	18	18	
Ash	38.6	38.6	38.6	38.6	38.6	35.5	35.5	35.5	35.5	35.5	35.5	
Carbon Conversion												
Solid Accountability	80	81	88	89	87	88	84	90	78	82	84	
Gas Make	71	68	73	85	80	83	81	61	78	74	80	
H2S concentration (ppm)	595	504	570	700	850	660	504	398	1318	1292	1246	
sulfur Retention (%)	46	57	48	34	39	45	49	69	33	38	40	
1-Lig, 2-Sub, 3-Bit, 4-Pet												
TC416 thermocouple (402-G)	857	869	890	895	888	876	868	872	776	771	839	
Calculate or Specify BEDRATE, *C* or *S*	S	S	S	S	S	S	S	S	S	S	S	
If Specified, Multiplier for Recirculation Rate	1.40	1.40	1.40	1.40	1.40	1.40	1.40	1.40	1.40	1.40	1.40	

Balanced Product - Product Gas) / (Product Gas) [Dry]	-3%	-1%	-1%	4%	5%	1%	-3%	-0%	-1%	5%	1%	
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SUMMARY OF THE [RAW]												
DRY PRODUCT GAS												
TOTAL PRODUCT GAS, scf/hr ----->	22898	23757	24730	23764	19358	23511	24476	24697	13557	13294	14180	
BEFORE	H2	5.4%	5.5%	5.6%	5.5%	8.6%	6.6%	5.6%	4.4%	13.1%	11.4%	13.4%
	CO	2.7%	3.0%	3.3%	3.4%	4.5%	3.7%	2.8%	2.4%	4.2%	3.6%	5.1%
	CH4	0.8%	0.8%	0.7%	0.7%	1.1%	0.9%	0.7%	0.5%	2.1%	1.8%	1.8%
BALANCING	CO2	12.9%	13.2%	12.8%	14.9%	17.8%	13.6%	11.7%	11.6%	21.3%	22.1%	22.6%
	N2	75.0%	75.0%	75.0%	75.0%	65.0%	75.0%	75.0%	75.0%	60.0%	60.0%	60.0%
	H2S	895	504	570	700	850	660	504	398	1318	1292	1246

SUMMARY OF THE [BALANCED]												
DRY PRODUCT GAS												
TOTAL PRODUCT GAS, scf/hr ----->	22308	23583	24367	24665	20304	23799	23754	24597	13449	13927	14337	
AFTER	H2	5.4%	5.5%	5.6%	5.5%	8.6%	6.6%	5.6%	4.4%	13.1%	11.4%	13.4%
	CO	2.7%	3.0%	3.3%	3.4%	4.5%	3.7%	2.8%	2.4%	4.2%	3.6%	5.1%
	CH4	0.8%	0.8%	0.7%	0.7%	1.1%	0.9%	0.7%	0.5%	2.1%	1.8%	1.8%
BALANCING	CO2	12.9%	13.2%	12.8%	14.9%	17.8%	13.6%	11.7%	11.6%	21.3%	22.1%	22.6%
	N2	75.0%	75.0%	75.0%	75.0%	65.0%	75.0%	75.0%	75.0%	60.0%	60.0%	60.0%
	H2S	895	504	570	700	850	660	504	398	1318	1292	1246

SUMMARY OF THE [RAW]												
Elemental and Mass Closure												
BEFORE	#MOLES H	-4%	-5%	-1%	2%	1%	0%	-9%	11%	14%	10%	11%
	#MOLES C	10%	18%	17%	8%	10%	10%	-7%	30%	7%	14%	14%
	#MOLES O	-12%	-13%	-8%	-13%	-10%	-4%	-5%	4%	5%	4%	8%
	#MOLES S	2%	5%	4%	4%	5%	3%	-3%	7%	1%	6%	4%
BALANCING	#MOLES N	-3%	-1%	-1%	4%	5%	1%	-3%	0%	-1%	5%	1%
	#MOLES Ca	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%
	#MOLES Mg	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%	0%
	MASS, %	-2%	0%	-1%	4%	4%	1%	-3%	0%	0%	3%	1%

SUMMARY OF CARBON REMOVALS

Fast Kinetics

% CARBON LOST BY GASIFICATION (Total)	3%	0%	0%	0%	0%	0%	0%	0%	0%	0%	5%	
% CARBON LOST BY COMBUSTION (Total)	76%	67%	81%	67%	81%	86%	81%	85%	46%	74%	80%	
% Carbon Removed In Filter Ash (Total)	19%	31%	19%	33%	19%	14%	19%	15%	54%	26%	15%	
% Carbon Removed In LASH (Total)	2%	2%	0%	0%	0%	0%	0%	0%	0%	0%	0%	
Char Carbon Accounted For (Comb + S Gasif + Filter + LASH)	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	
% Coal Carbon From Volatiles (Basis: Recycle Rate, Ultimate Analysis, 4x)	-6%	52%	34%	66%	30%	14%	6%	31%	89%	29%	-4%	

Single-Pass CHAR Carbon Conversion (Gasification)	5.7%	6.3%	23.1%	31.8%	14.5%	20.8%	16.4%	22.2%	10.0%	15.0%	9.9%	
Single-Pass CHAR Carbon Conversion (Combustion)	63%	100%	100%	100%	100%	100%	100%	100%	100%	100%	61%	
Bed Recycle Rate from Comb Zone nrg-Balance, lb/hr	9079	10528	4595	3647	9038	5282	6086	4673	3424	2365	6474	

Slow Kinetics - Mann et. al.

% CARBON LOST BY GASIFICATION (Total)	1%	0%	0%	0%	0%	0%	0%	0%	0%	0%	3%	
% CARBON LOST BY COMBUSTION (Total)	77%	67%	81%	67%	81%	86%	81%	85%	46%	74%	82%	
% Carbon Removed In Filter Ash (Total)	19%	31%	19%	33%	19%	14%	19%	15%	54%	26%	15%	
% Carbon Removed In LASH (Total)	2%	2%	0%	0%	0%	0%	0%	0%	0%	0%	0%	
Char Carbon Accounted For (Comb + S Gasif + Filter + LASH)	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	100%	
% Coal Carbon From Volatiles (Basis: Recycle Rate, Ultimate Analysis, 4x)	-4%	52%	34%	66%	30%	14%	6%	31%	89%	29%	-2%	

Single-Pass CHAR Carbon Conversion (Gasification)	3.1%	2.3%	4.2%	5.2%	2.3%	3.4%	3.8%	6.6%	13.9%	25.0%	6.4%	
Single-Pass CHAR Carbon Conversion (Combustion)	63%	100%	100%	100%	100%	100%	100%	100%	100%	100%	61%	
Bed Recycle Rate from Comb Zone nrg-Balance, lb/hr	9079	10528	4595	3647	9038	5282	6086	4673	3424	2365	6474	

[H2O] (Inlet)	15.5%	14.7%	14.2%	14.4%	21.4%	14.5%	14.1%	13.4%	40.3%	38.9%	38.3%	
[H2O] After Combustion (No WGS)	14.5%	13.6%	13.1%	13.2%	19.4%	13.4%	13.1%	12.4%	37.5%	36.0%	34.8%	
[H2O] After Combustion (WGS)	9.3%	8.5%	7.9%	7.9%	12.4%	8.2%	8.1%	7.4%	29.1%	27.3%	25.4%	
[H2O] After Stearns Gasification (WGS)	3.1%	8.2%	7.6%	7.7%	12.0%	7.9%	7.8%	7.2%	28.6%	26.8%	3.8%	

SUMMARY OF THE

SPREADSHEET RESULTS											
Carbon Conversion (Coal IN - Filter/LASH) / (Coal IN)	78%	84%	88%	89%	87%	88%	82%	90%	78%	82%	85%
Carbon Conversion (Coal IN - I1 - Balanced Product) / (Coal IN)	66%	65%	69%	84%	80%	78%	86%	59%	70%	71%	71%
% Sulfur On Sorbent CHAR & ASH (S IN - H2S) / (S IN) (m)	46%	56%	47%	31%	34%	44%	43%	68%	33%	34%	41%
Heat Loss from Mass & Energy Balances (H IN - H Out) / (LHVm)	-12%	-11%	-11%	-11%	-12%	-9%	-14%	-9%	-9%	-13%	-16%
Heat Loss from TRDU (Btu/hr)	-3.5E+005	-3.5E+005	-3.4E+005	-3.1E+005	-3.7E+005	-2.6E+005	-3.1E+005	-3.0E+005	-2.6E+005	-3.6E+005	-4.7E+005
LHV of Product Gas, 60 F w H2S w/o tar, Btu/scf	36	37	37	36	55	43	36	29	77	68	76
HHV of Product Gas, 60 F, w H2S w/o tar, Btu/scf (Dry Corrected)	78	76	75	73	136	88	81	63	235	212	220
LHV of Product Gas, 60 F, w H2S w/o tar, Btu/scf (Wet)	30	31	32	33	47	37	30	24	51	45	49
LHV of Product Gas, 60 F, w H2S w/o tar, Btu/scf (Wet Corrected)	56	57	56	59	98	65	57	43	92	82	85
Gasifier Cold Gas Efficiency, % of coal HHV	25%	25%	27%	29%	33%	34%	36%	21%	37%	32%	35%
Bed Recycle Rate from Comb Zone nrg-Balance, lb/hr	9079	10528	4595	3647	9038	5282	6086	4673	3424	2365	6474
ADJUSTMENTS TO STREAM FLOWS AND TROUBLE SHOOTING											
X, g Sorbent (Ca - S IN) / (S IN) (molar)	33%	33%	33%	33%	33%	26%	26%	26%	26%	26%	26%
If No then Sorbent is not in Excess											
Fraction of Carbon in LASH (Original)	2.5%	0.6%	2.2%	1.1%	1.2%	2.4%	1.7%	2.8%	0.9%	3.8%	3.8%
Fraction of Carbon in LASH (Modified)	2.8%	0.6%	2.2%	1.1%	1.2%	2.4%	1.7%	2.5%	0.9%	3.8%	3.8%
If No then there is not enough carbon in the Bed											
CO2/CO Mass Ratio Leaving Combustion Zone (No WGS)	0.5	0.5	0.4	0.3	0.3	0.4	0.4	0.4	0.6	0.6	0.5
CO2/CO Mass Ratio Leaving Combustion Zone (WGS)	1.9	1.7	1.3	1.3	1.5	1.4	1.5	1.3	4.0	3.6	2.7
(Balanced Product - Product Gas) / (Product Gas) (Dry)	-3%	-1%	-1%	4%	5%	1%	-3%	-0%	-1%	5%	1%
If No then Product gas has been adjusted too much											
(Balanced H2O - H2O) / (H2O) (Quenched)	-22%	-27%	-18%	-38%	-31%	-13%	-8%	9%	12%	2%	17%
If No then the water has been adjusted too much											
(Balanced Carbon Conversion) - ("UN-Balanced" Carbon Conversion)	-2%	-1%	-1%	3%	4%	1%	-3%	-1%	-1%	3%	0%
If No then the Carbon Conversion has changed dramatically											
Sum of Spreadsheet Checkers Must be Zero ---->	1	1	0	1	1	0	0	0	0	0	0
If No then there is error in the spreadsheet and debugging is required											
Kp - Kp Inventory Check Must be Zero ---->	0	0	0	0	0	0	0	0	0	0	0
Arrhenius "A"	32,500	32,500	32,500	32,500	32,500	32,500	32,500	32,500	32,500	32,500	32,500
Arrhenius "E"	31,300	31,300	31,300	31,300	31,300	31,300	31,300	31,300	31,300	31,300	31,300

COMPARISON OF CERTAIN SPREADSHEET RESULTS											
delta T (°C)	22	27	46	53	58	60	60	62	80	89	56
Average Mixing Zone T (°C)	879	896	936	948	946	936	928	934	856	860	895
Average Low T (°C)	857	869	890	895	888	876	868	872	776	771	839
Air in (scfh)	13,501	14,453	13,161	14,838	8,999	14,203	13,842	15,112	0	0	0
Oxygen in (scfh)	0	0	0	0	1,194	0	0	0	2,247	2,529	2,951
Nitrogen in (scfh)	5,149	5,249	5,312	5,395	5,020	5,196	6,204	6,238	6,552	6,985	6,894
Product Gas (scfh)	22,898	23,757	24,730	23,764	19,358	23,511	24,476	24,697	13,557	13,294	14,180
Coal Feed Rate (lb/hr)	419	465	448	416	451	441	382	505	416	433	461
Air Flow (lb/hr)	1033	1106	1160	1135	689	1087	1059	1156	0	0	0
Steam Flow (lb/hr)	175	172	172	172	215	168	168	168	327	330	332
Air/Coal Ratio (lb/lb)	2.57	2.48	2.70	2.84	1.59	2.57	2.89	2.39	0.00	0.00	0.00
Steam/Coal Ratio (lb/lb)	0.44	0.39	0.40	0.43	0.50	0.40	0.46	0.35	0.82	0.79	0.75
O2/Coal Ratio (lb/lb)	0.60	0.57	0.63	0.66	0.60	0.59	0.67	0.55	0.48	0.51	0.56
Fraction Carbon in Coal (Swan's)	0.44	0.44	0.44	0.44	0.44	0.41	0.41	0.41	0.41	0.41	0.41
Bed Recycle Rate, lb/hr (Everett's)	9,079	10,528	4,595	3,647	9,038	5,282	6,086	4,673	3,424	2,365	6,474
Bed Recycle Rate, lb/hr (Swan's)	6,485	7,520	3,282	2,605	6,456	3,773	4,347	3,338	2,446	1,689	4,624

APPENDIX B

RESULTS FROM EERC HAZARDOUS WASTE TREATABILITY STUDY

January 12, 2001

Mr. Curt Erickson, Program Manager
North Dakota Department of Health
Division of Waste Management
PO Box 5520
1200 Missouri Avenue, Room 302
Bismarck, ND 58506-5520

Dear Mr. Erickson:

Subject: EERC Report on EERC Hazardous Waste Treatability Study

Based upon previous conversations and correspondence with the North Dakota Department of Health, a hazardous waste treatability study was performed on waste generated at the University of North Dakota (UND) Energy & Environmental Research Center (EERC). This study was completed November 28, 2000. The enclosed report provides the details of the study successfully performed at the EERC within the guidelines of Section 33-24-02-04.6 of the North Dakota Administrative Code. As follows Subsection i, a report on the study is enclosed. As stated in Subsection j, representative samples of all waste materials generated by this study were analyzed and determined to be not hazardous waste. The City of Grand Forks has been contacted and has agreed to accept the waste at the landfill.

If there are any questions, please feel free to contact me at (701) 777-5172. You can also fax me at (701) 777-5181 or e-mail at dhajicek@undeerc.org. Thank you.

Sincerely,

Douglas R. Hajicek, PE
Manager, Advanced Power Systems

DRH/drh

Enclosure

c: John Hendrikson, EERC
Mike Swanson, EERC
Ken Grohs, EERC

Subsection i

Name: University of North Dakota (UND) Energy & Environmental Research Center (EERC)
Mailing Address: PO Box 9018
Grand Forks, ND 58202-9018
Shipping Address: 15 North 23rd Street
Grand Forks, ND 58203
Identification Number: NDD000819227

TREATABILITY STUDIES CONDUCTED

Materials were generated by the Transport Reactor Development Unit (TRDU), a research gasification system located at the Energy & Environmental Research Center (EERC) during Test No. PO63 conducted in September 1999. A brief description of the TRDU is contained in Appendix A. The Illinois No. 6 bituminous coal used during this test resulted in the generation of solids with high sulfide concentrations. The product analyses are shown in Table 1. During this test, there was 824 lb of bed material, 1596 lb of dipleg material, and 7313 lb of filter vessel material collected. Based upon the composite samples obtained from each of these product streams, they each, respectively, had sulfide concentrations of 4980 $\mu\text{g/g}$, 3530 $\mu\text{g/g}$, and 2650 $\mu\text{g/g}$.

TABLE 1

TRDU Product Analyses			
	Bed		Filter
	Material	Dip Leg	Vessel
Proximate Analysis, wt%			
Moisture	0.0	0.1	0.4
Volatile Matter	2.7	2.1	7.0
Fixed Carbon (ind.)	0.0	0.0	20.8
Ash	97.3	97.8	71.8
Heating Value, Btu/lb	0	864	5473
Sulfides, $\mu\text{g/g}$	4980	3530	2650

Treatment of these materials was in the EERC circulating fluidized-bed (CFB) combustor. The CFB is a research combustion system, and a brief description is contained in Appendix B. A CFB-type system would be well suited for the economical and environmental disposal of the solid product streams that would be generated by a commercial transport gasification system. This solid fuel by-product is a low-Btu small-sized material that would be difficult to burn in a pulverized-coal-fired system and could only be burnt in a bubbling fluidized bed at very low velocities, thus requiring a much larger system compared to a CFB combustor. Operational CFB

combustion data will allow for a more optimized design of a commercial CFB system. Steam or steam and electrical power would be produced by the combustion of these solids.

The first test conducted on November 19–20, 2000, was terminated early because of operational difficulties encountered. A normal start-up procedure was used, with initial heatup on natural gas to about 800°F and then switching over to coal for final heatup to about 1550°F. The more reactive Wyodak subbituminous coal was needed to heat up the system to full operational temperatures, because of limitations with the natural gas preheat system, before introducing the relatively low-Btu test fuel that would not likely ignite at the low temperatures. Several indications pointed to the lack of good bed material circulation being established even at the start of this test. The most likely explanation is that the silica sand bed material (~600- μ m average size) that was selected for use in this CFB test is suspected to be slightly oversized for this combination of fluidizing velocity and particularly for the extremely fine size of this fuel. Normal fuel feed size for the CFB is minus 1/4 inch (6.35 mm), while this material was mostly less than 0.1 mm. The unexpected lack of good circulation resulted in the majority of the fuel being burned at the top of the combustor while the lower portions of the combustor continually dropped off in temperature. Several corrective adjustments were made (by redistributing the combustion air into the combustor and external heat exchanger), but a recovery could not be achieved. As a result of high temperatures, there were deposits formed at the entrance into the primary cyclone and also at the bottom of the primary cyclone where solids normally flow down the downcomer into the external heat exchanger and then back to the combustor. The system was shut down and allowed to cool down for cleaning and a subsequent restart.

A much smaller than normal silica sand (~150 μ m average size) was selected for the restart. There was some concern about being able to retain this size material in the system, but no problems with excessive carryover into the baghouse were encountered. The operational conditions for both tests are shown in Table 2. The second test was divided into three test periods based upon average combustor temperatures. It is suspected that during Period 2 more of the fuel feed material was from the lower-Btu TRDU bed material and dipleg material than the filter vessel material. This did result in lower average bed temperatures (~1400°F) and required a minimal addition of the start-up coal (~5 lb/hr) to help ensure that sufficient bed temperatures could be maintained to successfully oxidize the sulfides present in the feed material. The average oxygen content of the flue gas during Test Period 2 was 10% or greater. Test Periods 1 and 3 are very similar, with average bed temperatures of 1520°F and oxygen content at about 2% to 3%. The overall fuel feed rate for Test Periods 1–3 was 402 lb/hr. Test conditions for the second test are shown in Figures 1–3. Figures 1 and 2 show combustor and downcomer and external heat exchanger temperatures, respectively. Figure 3 shows sulfur dioxide emissions obtained during Test 2.

There was no limestone addition for sulfur capture during any of the testing. Based upon visual observations, the sulfur dioxide emissions were somewhat greater than 1000 ppm and much less than 5000 ppm for the November 20 test during the brief, less than 45 minutes, high-temperature portion of this test. Levels greater than 1000 ppm were not set to be automatically

TABLE 2

CFB Operating Conditions				
		Period 1	Period 2	Period 3
Date	11/20	11/27–28	11/28	11/28
Start Time	09:45 AM	11:00 PM	08:30 AM	02:30 PM
Stop Time	10:30 AM	07:30 AM	02:00 PM	10:00 PM
Average Temperatures, °F				
Combustor	1714	1519	1402	1520
Downcomer	1230	1525	1402	1519
External Heat Exchanger	707	1491	1382	1529
Combustion Air, scfm	309	406	416	405
Fuel Feed Rate, lb/hr	216	451	396	361
Velocities, ft/sec				
Combustor	9.7	13.0	12.6	13.0
External Heat Exchanger	2.4	2.0	2.0	2.0
Gas Emissions				
Oxygen, %	3.7	1.9	10.0	3.1
Nitrogen oxides, ppm	430	48	99	69
Sulfur Dioxide, ppm	>1000	645	52	437
Sulfur Dioxide, lb/hr	>3.6	2.8	0.3	2.1
Solid Emissions, lb/hr				
Bed Material	0	21	173	40
Secondary Cyclone	0	89	113	259
Baghouse	61	51	40	45

recorded by the data acquisition system. The system can be manually switched over to the 5000 ppm range if required. It had been planned to add limestone if sulfur dioxide emissions exceeded 5000 ppm for any sustained periods. For the flue gas flow rate at which this test was conducted, a sulfur dioxide emission of 5000 ppm would produce 18.2 lb/hr of sulfur dioxide, less than the 18.3 lb/hr limit for the EERC. Solids addition and removal rates are shown in Tables 3 and 4, respectively.

Representative samples were obtained for the November 20 test, from the low-temperature Test Period 2, and from composites of Test Periods 1 and 3. Test results for the sulfide content of these sample are shown in Appendix C. All samples were successfully treated, resulting in sulfide contents of less than 0.2 µg/g. Based upon these results, all of the treated materials will be disposed of in the Grand Forks city landfill.

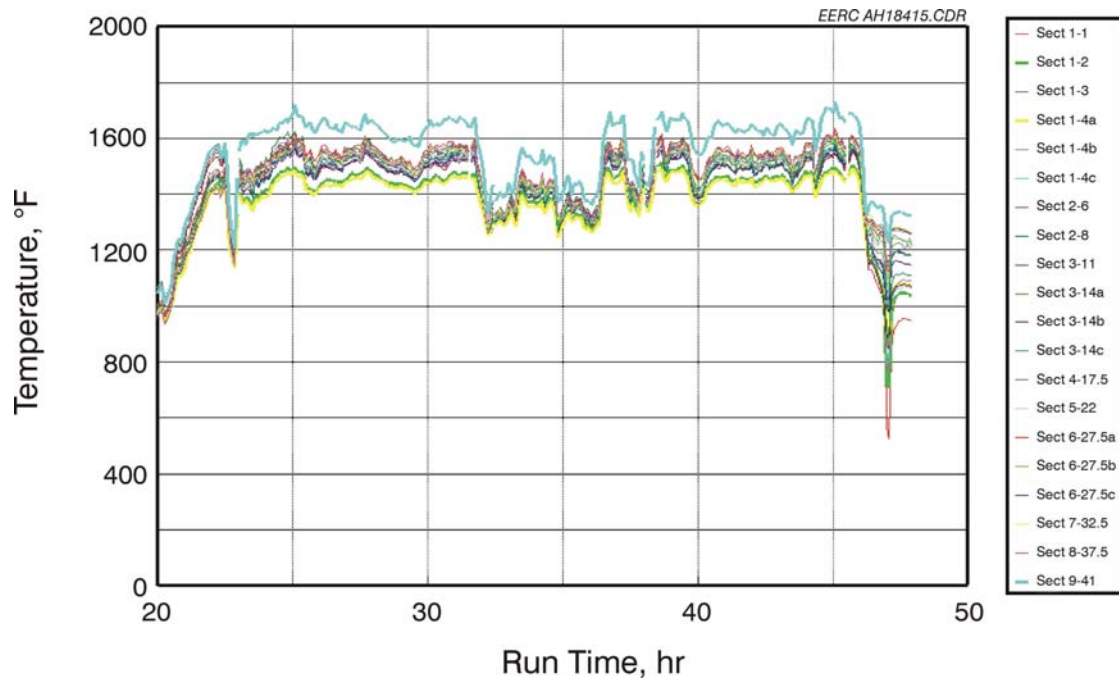


Figure 1. Combustor and downcomer temperatures.

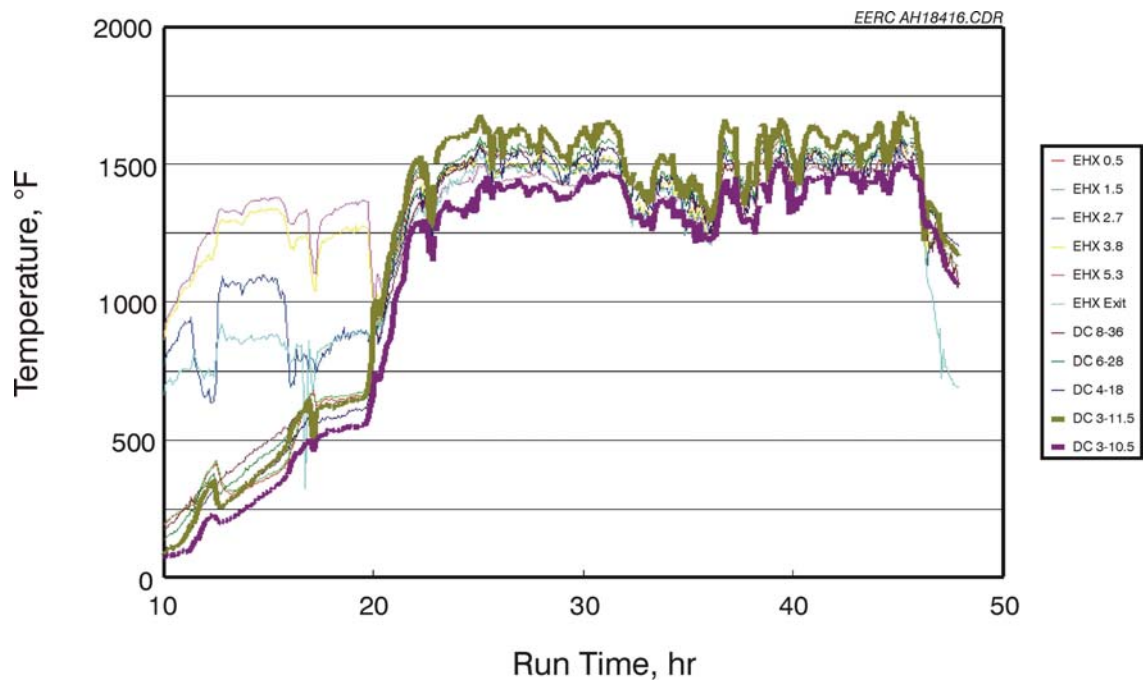


Figure 2. External heat exchanger temperatures.

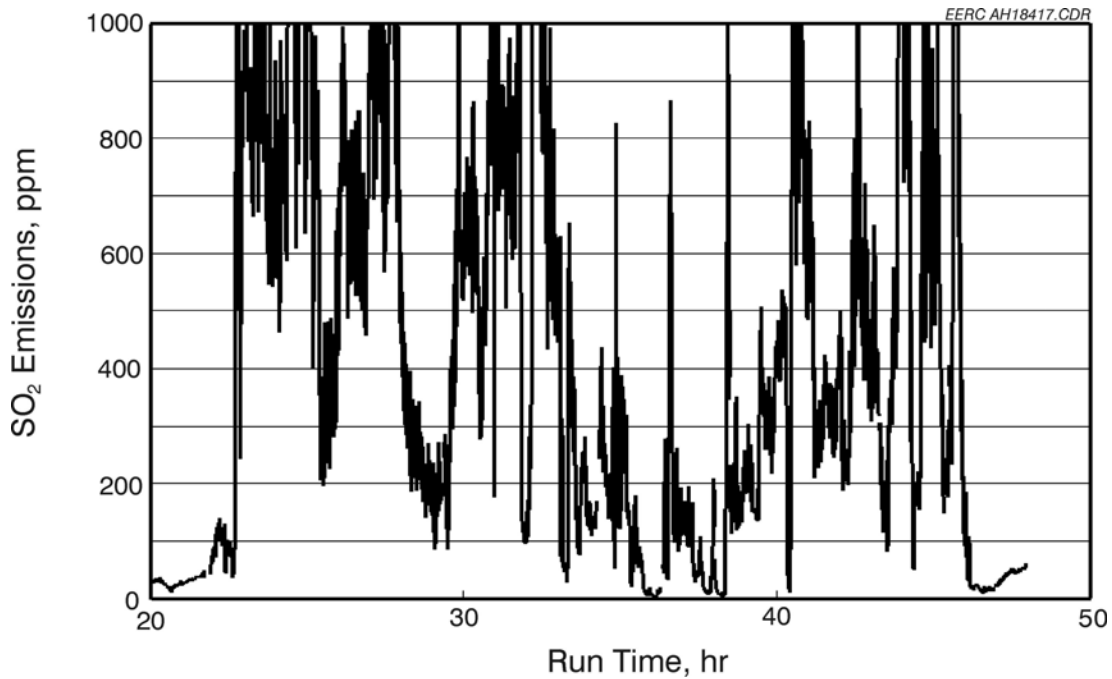


Figure 3. Sulfur dioxide emissions obtained during Test 2.

TABLE 3

CFB Solids Addition

Fuel Feed Rate					
Start Date	Start Time	Stop Time	Hopper Net, lb	Hopper Feed Rate, lb/hr	Period Feed Rate, lb/hr
11/20/1900	09:45 AM	10:30 AM	162	216	216
Period 1					
11/27/1900	11:00 PM	01:50 AM	2080	734	
11/28/1900	01:50 AM	07:15 AM	1640	303	451
Period 2					
11/28/1900	07:15 AM	01:10 PM	2344	396	396
Period 3					
11/28/1900	01:10 PM	07:15 PM	2352	387	
11/28/1900	07:15 PM	10:00 PM	836	304	361
Totals/Average			9252	402	402

Bed Material Added (Silica Sand)

Date	Amount, lb
11/20/1900	1200
11/27/1900	800
Total	2000

TABLE 4

CFB Solids Removal			
Bed Material Removed			
Date	Barrel	Drain Location	Weight, lb
11/20/1900	1	Combustor	635
11/20/1900	2	Combustor	310
11/20/1900	1	Downcomer	326
Total			1271
11/28/1900	1	Combustor	483
11/28/1900	2	Combustor	508
11/28/1900	3	Combustor	496
11/28/1900	4	Combustor	416
11/28/1900	1	Downcomer	668
Total			2571
Secondary Cyclone Ash Material Captured			
Date	Barrel	Drain Location	Weight, lb
11/28/1900	1	Cyclone	528
11/28/1900	2	Cyclone	380
11/28/1900	3	Cyclone	408
11/28/1900	4	Cyclone	526
11/28/1900	5	Cyclone	530
11/28/1900	6	Cyclone	637
Total			3009
Baghouse Ash Material Captured			
Date	Barrel	Drain Location	Weight, lb
11/20/1900	1	Baghouse	46
11/28/1900	1	Baghouse	208
11/28/1900	2	Baghouse	229
11/28/1900	3	Baghouse	219
11/28/1900	4	Baghouse	166
11/28/1900	5	Baghouse	236
Total			1058
Total Solid Material Collected			Weight, lb
11/20/1900			1317
11/28/1900			6638
Total			7955

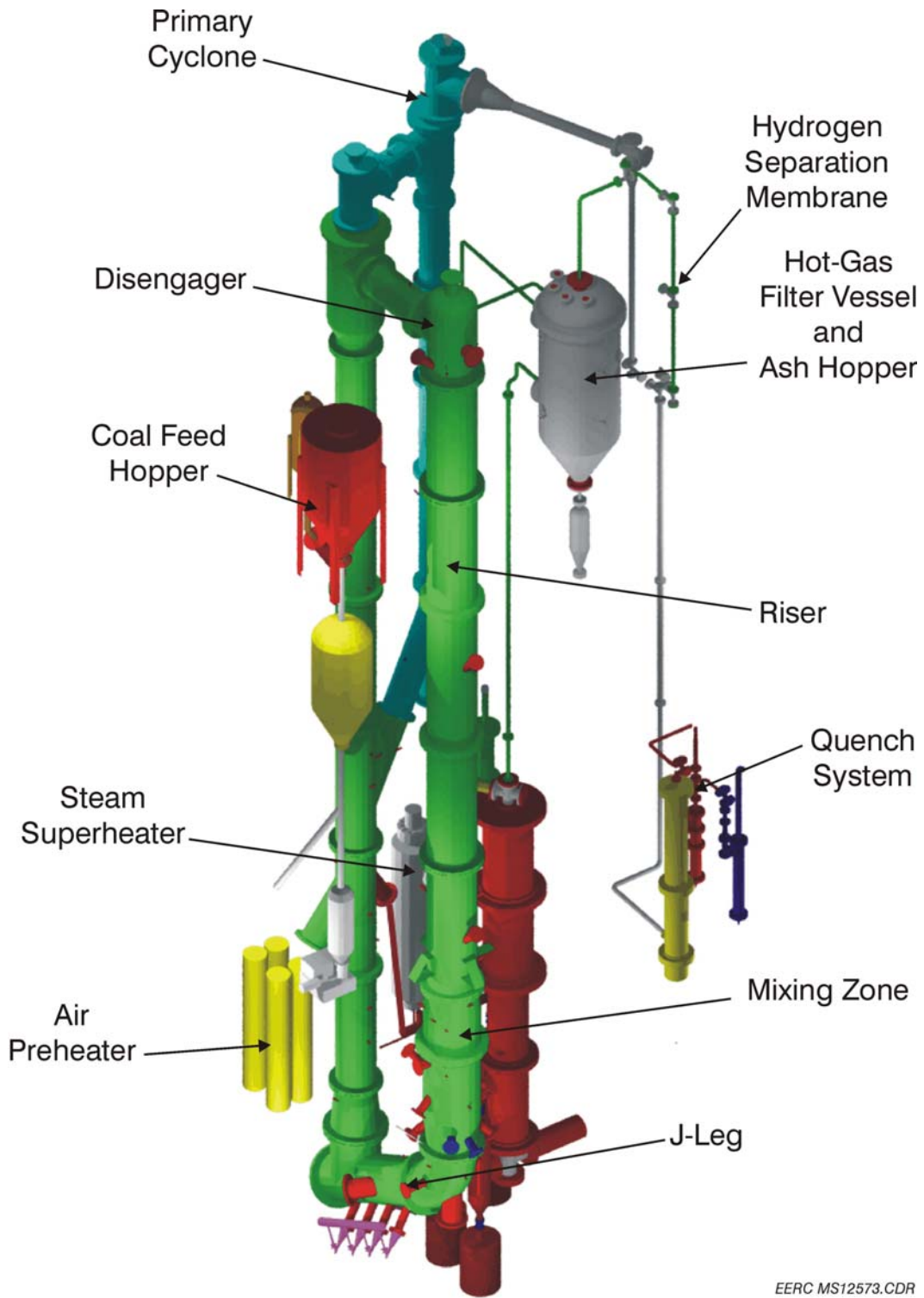
In conclusion, a CFB system is well suited for the successful conversion of this type of material into a nonhazardous waste for disposal in both an economic and environmental manner. This test provided a couple of basic data points demonstrating the CFB's potential for use in this process. Detailed design data for an optimized design would require a much more complete test matrix. Test variables would include parameters like bed material size, fluidizing velocities, average combustor bed temperatures, excess air, and limestone addition for better sulfur capture. No further testing is planned at this time. Additional testing is dependent upon approval from the North Dakota Department of Health and obtaining additional funding for this type of research. Use of the EERC CFB system for conversion of high sulfide materials to nonhazardous waste will be normally more expensive than shipping to an off-site licensed disposal facility. This is due to a number of reasons. Usually only a relatively small amount of material is being disposed of, which only allows the use of a research facility for a short period of time. A thorough research program would require much more extensive characterization of the feed and product streams as compared to simple disposal.

APPENDIX A

THE EERC TRANSPORT REACTOR DEVELOPMENT UNIT

The transport reactor development unit (TRDU) is a 200–300-lb/hr pressurized circulating fluid-bed gasifier. The TRDU has an exit gas temperature of up to 2000°F, a gas flow rate of up to 350 scfm, and an operating pressure of 120 psig. The TRDU system can be divided into three sections: the coal feed section, the TRDU, and the product recovery section. The TRDU proper, as shown in Figure A-1, consists of a riser reactor with an expanded mixing zone at the bottom, a disengager, and a primary cyclone and standpipe. The standpipe is connected to the mixing section of the riser by a J-leg transfer line. All of the components in the system are refractory-lined and designed mechanically for 150 psig and an internal temperature of 2000°F.

The hot-gas filter vessel (HGFV) is designed to handle all of the gas flow from the TRDU at its nominal operating conditions. This vessel has a 48-in. inner diameter and is 185 in. long, with a refractory inside diameter of 28 in. and a shroud diameter of 24 in. Filter vessel design capabilities include operation at elevated temperatures (to 1750°F) and pressures (up to 150 psig), with the initial test program operating in the 1000°–1200°F range. The HGFV can operate with filter face velocities in the range of 2.5 to 10 ft/min. Up to nineteen 1.5-meter candles can be installed in the filter vessel. An existing heat exchanger limits the current hot-gas filter system to operation between 800° and 1200°F. An unheated nitrogen backpulse system was constructed to test the effects of backpulsing parameters on candle performance and cleanability. The nitrogen backpulse system was constructed to backpulse up to four sets of four- or five-candle filters in a time-controlled or differential pressure-controlled sequence.



EERC MS12573.CDR

Figure A-1. TRDU with HGFV in EERC gasification tower.

APPENDIX B

DESCRIPTION OF THE CIRCULATING FLUIDIZED-BED COMBUSTION SYSTEM

A schematic of the overall circulating fluidized-bed combustion (CFBC) system is shown in Figure B-1. The overall system is divided up into the following subsystems:

- Combustion Air System
- Flue Gas System
- Flue Gas Recirculation System
- Ash-Fouling Section
- Fuel and Sorbent System
- Combustor
- Solids Recirculation System
- Natural Gas-Fired Preheater
- Combustor Heat Exchange System
- External Heat Exchange System
- Flue Gas Cooling Water System

A forced-draft blower supplies combustion air and secondary air to the combustor. The combustion air heat exchanger is a shell and tube heat exchanger that uses hot flue gas to preheat the combustion air before it enters the combustor. Total combustion air flow is controlled by the amount of bypass through the combustion air bypass valve located directly after the combustion air heat exchanger. The secondary combustion air control valve determines the ratio of combustion air which enters the test furnace above the distributor plate to the amount of

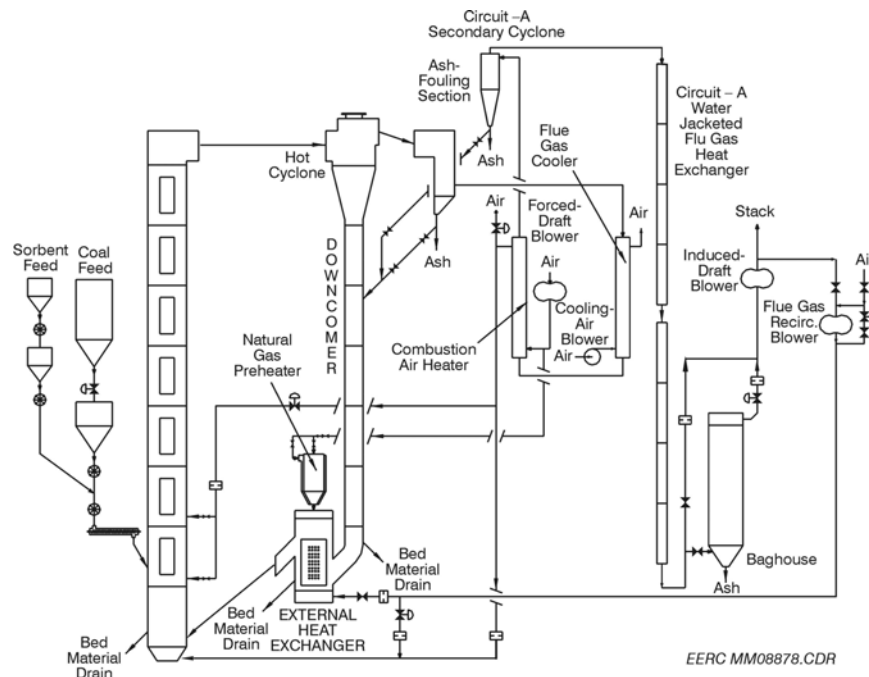


Figure B-1. Schematic of CFB pilot plant.

combustion air introduced into the combustor plenum below the distributor plate. The secondary combustion air can be introduced through manifolds at two different levels, located 5' 9" and 10' 6" above the distributor plate in Sections 2 and 3, respectively, of the combustor. Four 3-inch manual gate valves at each level are used to select where overfire air is introduced into the combustor.

Flue gas exits the top of the combustor, then flows through a refractory-lined primary cyclone with an inside diameter of 25 inches, the ash-fouling section, an air-cooled flue gas cooler, the combustion air heater, an 18-inch stainless steel secondary cyclone, eight water-jacketed flue gas heat exchangers, and through either the flue gas bypass or the baghouse. Temperatures and pressures are monitored throughout the flue gas system. Flue gas is drawn through the induced-draft (ID) blower where it finally enters a stack for release to the atmosphere. Flue gas flow is controlled by the amount of air allowed into the ID blower through the ID fan bypass valve. The ID fan bypass valve is computer-controlled and continually adjusted to maintain -2-inch pressure at the inlet of the primary cyclone.

The flue gas recirculation blower is used to supply either air or flue gas to the external heat exchanger (EHX) and to supply flue gas to the combustor for flue gas recirculation testing. Manual gate valves located upstream of the blower allow either air or flue gas to enter the blower.

Primary and secondary combustion air, flue gas recirculation, and flue gas flow rates are measured using orifice plates. Instrumentation is interfaced with the data acquisition/control system to record and display the flow rates. Orifice differential and static pressures are also monitored with magnehelic pressure gages.

The ash-fouling section is located at the exit of the 25-inch primary cyclone. Two air-cooled stainless steel probes maintained at 1000°F are present in the ash-fouling section to detect potential ash deposition or slagging. A hopper attached to the bottom of the ash-fouling section is connected to the downcomer via a drain leg containing two pneumatically actuated gate valves for ash recirculation. Three pneumatically actuated gate valves are used to allow the solids collected downstream by the secondary cyclone to be either routed back into the downcomer or to a collection barrel located on the ground floor. The length of time that any of these five pneumatic valves are open or closed is controlled with the data acquisition/control system.

The fuel storage hopper has a capacity of about 3000 pounds, which is transferred to a permanent feed hopper in approximately 600-pound increments. A gate valve is used to recharge the fuel feed hopper. The fuel feed hopper is suspended from a load cell; approximate fuel feed rates are calculated from the weight loss of the hopper over time. At the bottom of the weigh hopper, a rotary valve with an electronic speed controller is used to control the fuel feed rate.

The combustor is a series of refractory-lined sections bolted together. Each section has 2 inches of hard, abrasion-resistant refractory used in combination with 7 inches of insulating refractory. The bottom plenum section has the primary combustion air entrance and a bed material drain. The first combustor section (Section 1) has the solids recirculation return from the EHX. A removable stainless steel nozzle distributor plate is installed between the plenum and first combustor section. The next seven sections (Sections 2–8) each have two doorways on opposite sides for the installation of either blank refractory doors or heat exchanger panels. At

this time, twelve of the possible fourteen heat exchanger panels are installed in the combustor, two each in Sections 2, 3, 4, 7, and 8, and one each in Sections 5 and 6. Section 2 has the entrance for gravity feed of fuel and sorbent and the first set of secondary combustion air ports. Section 3 has the second set of four secondary combustion air ports. Section 9, the combustor exit, connects to the primary refractory-lined cyclone. Thermocouple and pressure taps are present in all of the combustor sections. All pressure taps are continuously purged with air to keep them open for accurate pressure measurements.

The refractory-lined components of the solids recirculation system include the primary cyclone, the downcomer, and the EHX. Solids that are captured by the primary cyclone drop into the downcomer and travel downward into the EHX. Thermocouples monitor the temperature at the entrance and exit of the primary cyclone. The EHX has a plenum section into which either air or flue gas can be introduced. A removable stainless steel distributor plate is installed between the plenum and the main body of the EHX. The natural gas-fired preheater, described later, is attached to the top section of the EHX. Sixteen U-shaped stainless steel water-cooled heat exchanger tubes are installed in a removable refractory-lined door in the EHX. Thermocouple and pressure taps are distributed along the sections of the downcomer and in the EHX.

The preheater combustion chamber is constructed with inner and outer stainless steel shells. The natural gas-fired burner is bolted to the top of the preheater and fires downward. To maintain an acceptable operational temperature on the inside surface of the preheater, air is circulated through a baffled cooling jacket. Cooling air enters at the top of the preheater and flows downward, where it combines with the combustion gases at the bottom of the preheater transition cone. Preheater combustion air and the cooling jacket air are supplied by the forced-draft (FD) blower. A butterfly valve in the 4-inch supply line from the FD blower and a gate valve between the preheater and the EHX isolate the system when it is not being used. There are butterfly valves in the combustion air and cooling air lines for control purposes. There are also orifice plates in each line with magnehelics to monitor the flow rates. The flow of natural gas to the main and pilot burners are controlled with flowmeters located in the control room. A flame safety system is located in the control room to shut off the flow of natural gas to the preheater if 1) a flame is not present in the preheater, 2) combustion air is not being supplied to the preheater or cooling jacket, or 3) the combustion air pressure is greater than the natural gas pressure supplied to the preheater.

The rate of water flow to the combustor heat exchangers (CHX) is measured individually for each door by flowmeters and controlled by globe valves installed above the flowmeters in the CHX panel boards. Total flow is measured with an in-line turbine flowmeter, which includes a bypass to allow for maintenance or repair during operation. An air system is connected to the inlet manifolds of each of the heat exchange panels. Air is used to cool the heat exchanger panels during operation prior to the introduction of water. Each inlet manifold has a selector switch to allow for the proper distribution of either air or water through the manifold into the heat exchanger tubes of the panels.

There are sixteen heat exchange coils installed in the external heat exchanger door. Each U-shaped heat exchanger is constructed out of 1-inch stainless steel pipe with ½-inch stainless steel tubing at each end. Each of eight circuits have a flowmeter and flow control valve mounted in a panel board to monitor and control the flow of water. Total flow is measured with an in-line turbine flowmeter, installed with a bypass to allow for maintenance or repair during operation.

Three different configurations are used: two using a single tube, four with two tubes in series, and two with three heat exchanger tubes connected in series. A thermocouple is located in the exit of each circuit to measure the water exit temperature.

APPENDIX C

COPIES OF ANALYSES AND DATA SHEETS

Final Results

November 1, 1999

Set Number: 49910

Request Date: Monday, November 01, 1999

Fund#: 4506

Due Date: Monday, November 15, 1999

PI: Mike Swanson

Set Description: Ash Samples for Sulfide

Contact Person: M. Swanson

Sample	49910-01
	49910-01 PO62 L ash hopper 7/13/99 time2200
	Sulfide 124 $\mu\text{g/g}$
	49910-02 PO63 Stand pipe 0645 8/31/99
	Sulfide 4980 $\mu\text{g/g}$
	49910-03 PO63 Dipleg 0645 8/31/99
	Sulfide 3530 $\mu\text{g/g}$
	49910-04 PO63 Filter 0645 8/31/99
	Sulfide 2650 $\mu\text{g/g}$

Final Results

November 4, 1999

Set Number: 49911

Request Date: Thursday, November 04, 1999

Fund#: 4506

Due Date: Thursday, November 18, 1999

PI: Mike Swanson

Set Description: Ash Samples for Sulfide

Contact Person: M. Swanson

Sample	49911-01
49911-01	PO63 Filter 0645 8/31/99
	Sulfide 1820 $\mu\text{g/g}$
49911-02	PO63 Filter 1335 8/30/99
	Sulfide 349 $\mu\text{g/g}$
49911-03	PO63 Standpipe 1335 8/30/99
	Sulfide 333 $\mu\text{g/g}$
49911-04	PO63 Standpipe 1505 8/31/99
	Sulfide 1660 $\mu\text{g/g}$
49911-05	PO63 Filter 1505 8/31/99 BBL#24
	Sulfide 425 $\mu\text{g/g}$

ANALYTICAL RESEARCH LAB - Final Results

December 4, 2000

Set Number: 50154

Request Date: Monday, December 04, 2000

Fund#: 4506

Due Date: Monday, December 18, 2000

PI: Mike Swanson

Set Description: Ash Samples for Sulfide

Contact Person: M. Swanson

Sample 50154-01

50154-01	11/20/00 BBL#1 Bed Material	
Sulfide		< 0.1 µg/g
50154-02	11/20/00 BBL#1 BH Ash	
Sulfide		< 0.2 µg/g
50154-03	11/28/00 Cyclone Ash Composite	
Sulfide		< 0.2 µg/g
50154-04	11/28/00 BH Ash Composite	
Sulfide		< 0.2 µg/g
50154-05	11/28/00 Bed Material Composite	
Sulfide		0.26 µg/g
50154-06	11/28/00 BBL#3 BH Ash (low temp)	
Sulfide		< 0.2 µg/g
50154-07	11/28/00 BBL#3 Cyclone Ash (low temp)	
Sulfide		< 0.2 µg/g

Run No. CFB-NDI-0100

BED MATERIAL (BED - 11270)

Bbl. No.	Test No.	Start Date	Start Time	Start Name	Stop Date	Stop Time	Stop Name	Gross (lbs)	Tare (lbs)	Net (lbs)
1		11-20			11-20	←				635
2		11-20			11-20	←				310
<hr/>										
1		11-28			11-28			532	49	483
2		11-28			11-28			556	48	508
3		11-28			11-28			546	50	496
4		11-28			11-28			466	50	416
<hr/>										
DOWN COMER										
1		11-20	←		11-20	←				326
<hr/>										
1		11-28			11-28			718	50	668