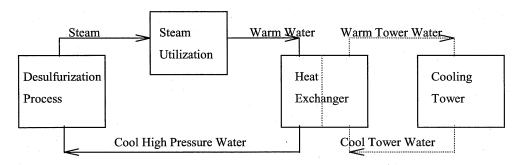
VI. ADDITIONAL PROCESS CONSIDERATIONS

1. Steam Generation

The coal gas desulfurization with sulfur production overall process is exothermic. DSRP and AHGP both require heat removal for condensation and to maintain reaction temperatures. The heat removal requirements create the opportunity to generate high pressure steam that could drive plant equipment or be incorporated into the plant's power generation steam cycle.

Steam generation has been modeled as a closed loop. Steam is generated by removing heat from the desulfurization process. The steam is then utilized, by undefined means, condensed, cooled and the condensate is reused. Cooling tower water is used to cool the steam-condensate loop (Figure 8). There are benefits to having a self-contained loop for steam production. First, it makes it easy to maintain steam-condensate purity, which reduces fouling and corrosion. It also allows for higher cool water feed temperatures (~90°C), which increases steam production.

Figure 8: Schematic for HGD Steam Generation



The steam generated from the HGD process was assumed to be at 950 psia and 441°C (Appendix F). Since desulfurization would be incorporated into a larger power generating plant, it is not possible to discern the most useful steam conditions without knowledge of the power generation facility. It is likely that steam generated from the HGD would be utilized by existing power plant equipment. Since the end use of the steam generated is unknown a generic dollar

credit for the steam generated was used for the economic analysis. Peters and Timmerhaus (1991) state that 500 psig steam was worth \$ 0.0039/lb in 1990; this value was used during the economic assessment. The benefit calculated should be a conservative value since the simulated steam produced is at a higher pressure (950 psia) and the economic calculations use 1996 as a basis. However, another source notes that for 900 psi and 441°C steam, 1 kWh power generation can be expected per 22.44 pounds of steam (Noyes). The economic credit from the conversion of steam to power according to this relationship was less than the credit obtained using the Peters and Timmerhaus relationship. Since the Peters and Timmerhaus credit value is conservative and still predicts a larger benefit, the Peters and Timmerhaus value was used.

2. Material of Construction

Type 310 stainless steel (25%Cr - 20%Ni) should be used for the construction of equipment that contacts sulfur species. Type 310 stainless steel (SS 310) will be more durable than type 316 stainless steel (SS 316) (17%Cr - 8% Ni - 2%Mo). Higher chromium content gives SS310 greater oxidation resistance, and the higher nickel concentration gives improved resistance to carburization (EPRI). Cost data for SS310 is not contained in ASPEN so SS316 material cost factors were used.

3. Sulfur Storage

Transporting molten sulfur is preferred over solid sulfur. Liquid sulfur is easier to transport and reduces handling losses. It will be necessary to store the molten sulfur before it is shipped out by train. The storage tank should be capable of storing several days worth of recovered sulfur. It should also be equipped with a heat exchanger to keep sulfur molten. The costs of the sulfur storage tanks were calculated using ASPEN assuming SS 310 was used to construct storage for seven days of sulfur production (SS 316 was entered in ASPEN due to lack of data for SS 310).

4. Process Operation

The DSRP should be the easier process to operate. Balancing the SO₂ production and consumption in the AHGP appears to be particularly difficult. The difficulty arises from the reaction of FeS with SO₂ to form elemental sulfur. The reaction's equilibrium varies significantly with temperature. If the reactants are too thermodynamically favored, less SO₂ will be consumed than expected. However, SO₂ production will remain constant (sorbent oxidation being driven to completion by oxygen). Thus, if the reaction:

$$3 SO_2 + 4 FeS \iff 7/2 S_2 + 2 Fe_2O_3$$
 (7)

does not reach design conversions, SO₂ flow will increase and sulfurized sorbent will be returned to the desulfurization reactor. With SO₂ already present in great excess the increased SO₂ flow will not significantly shift equilibrium towards the products.

It is recommended that the AHGP be operated at conditions that will cause a net consumption of SO₂. Replenishing depleted SO₂ levels can easily be accomplished by increasing the oxygen feed. Excess oxygen will convert elemental sulfur into SO₂.

Preventing the build up of impurities in the SO₂ loop contributes to the complexity of the AHGP. Venting a portion of the loop is undesirable since it contains mostly SO₂. Venting would release SO₂, emissions the system is designed to eliminate. Operating the AHGP requires determining the rate at which impurities build up in the recycle loop and the appropriate purge stream for the rate of build up. The purge stream should be fed to the desulfurization reactor, reducing the release of SO₂.

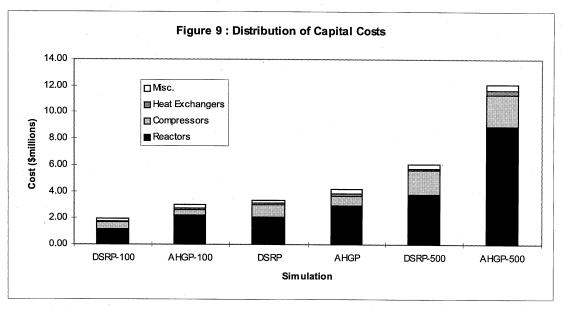
VII. ECONOMIC ANALYSIS

1. Capital Expenditures

The AHGP requires more capital investment than the DSRP. Reactors account for over half of the capital investment. The higher cost of AHGP reactors results in an higher overall capital investment necessary for the AHGP (Figure 9). The majority of equipment was costed using ASPEN. Equipment costed by ASPEN has a purchase date set at June, 1996. Equipment contacting sulfur will experience less corrosion when constructed of stainless steel 310 (SS310). Since ASPEN lacks material of construction correction factors for SS310, SS316 values were used. While the majority of equipment was costed using ASPEN, the equipment that comprises the majority of the capital expenditures, such as the reactors, were estimated by other means.

The reactor costs were calculated using a procedure outlined in Peters and Timmerhaus (1991). The reactor costs were determined using the amount of steel required for their construction. The procedure is described in appendix G, and the calculations are contained in appendix H and appendix I. The reactor cost includes the cost of installation.

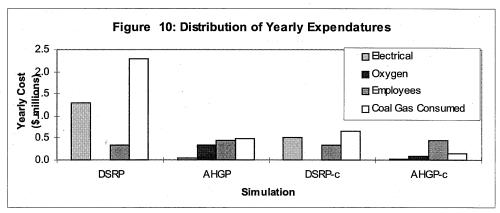
Another piece of equipment not costed by ASPEN is the PRESAIR - air compressor used in the DSRP. PRESAIR costs were determined by scaling a price quote for the Ingersoll-Rand Centac air compressor. The Centac Model 2CV23M3EEPF, capable of raising 2,250 acfm to 280 psia, was quoted at \$241,000. Extrapolation was used in determining the cost of compressors needed for the different flow rates. Figures in Peters and Timmerhaus (1990) were used to determine the rate at which compressor costs change with varying flow rates.

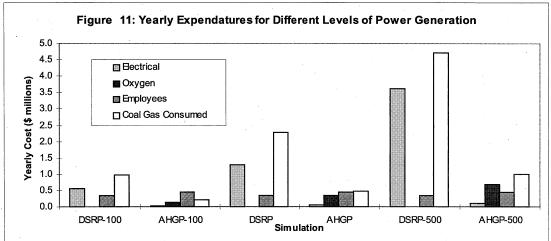


There are additional capital costs not included in this report, two of which, piping costs and sorbent/catalyst costs, will probably be significant. There will be other expenses, like additional office space for employees, which are site dependent. The site dependent expenses should not have an significant effect on the total capital investment calculations. At this stage of investigation the piping and sorbent/catalyst cost are assumed identical for both HGD process. If this assumption is valid than a comparison of the overall capital costs for the AHGP and the DSRP will not be affected by their absence.

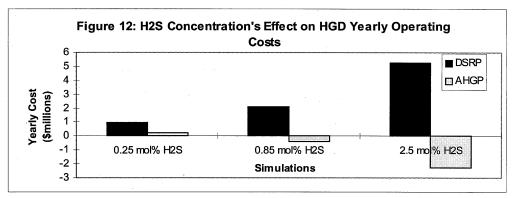
2. Yearly Operating Costs

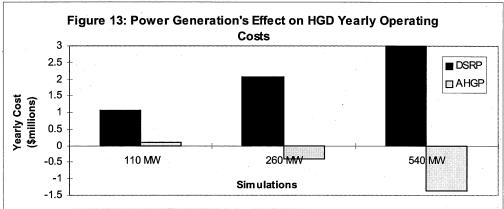
The AHGP has a lower yearly operating cost than the DSRP. Figures 10 and 11 show the distribution of the major yearly expenditures for both processes.



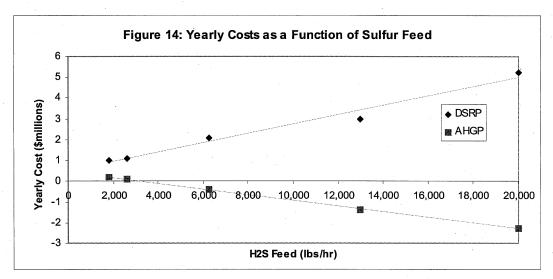


The bases cases (DSRP and AHGP) have coal gas feeds containing 0.85 mol% H₂S and produce 260 MW. Most of the yearly expenditures decline as the amount sulfur in the coal gas is decreased (DSRP-c and AHGP-c have feeds containing 0.25 mol% H₂S). The exception is the yearly costs of additional employees, which have been assumed to be dependent on the complexity of the HGD process and not its size. As the sulfur concentration decreases both the absolute expenditure difference (DSRP cost - AHGP cost) and the relative expenditure difference ([DSRP cost - AHGP cost] / AHGP cost) decrease. This decrease indicates that the competitive advantage of the AHGP is smaller for cleaning a coal gas stream containing a low H₂S concentration. The same trend exists comparing the economics of different levels of power generation: the AHGP's yearly economic advantage over the DSRP declines as the overall power generation is decreased.





In assessing the yearly cost of maintaining HGD, benefits of the process should also be accounted for. Two sources of credit were observed: the recovery of sulfur and the production of steam. Sulfur credits where consistently larger than steam credits within the same simulation. The sulfur credits remained virtually unchanged between corresponding DSRP and AHGP simulations. Figure 12 and 13 show that for several AHGP conditions the credits are larger than the expenditures. This results in negative yearly operating costs. When larger amounts of sulfur are removed, the yearly expenditures combined with the sulfur and steam credits result in negative yearly costs for the AHGP. In such cases it is more profitable to use the AHGP, then to leave the coal gas stream untreated (if Federal Regulations allowed). The profit that results from the sale of recovered sulfur (Appendix M) allows the AHGP to be more profitable than generating power without desulfurization.



The yearly costs have a linear dependence on the amount of sulfur being processed. This can be seen by comparing all simulations (DSRP, DSRP-b, DSRP-c, DSRP-100, DSRP-500, AHGP, AHGP-b, AHGP-c, AHGP-100, AHGP-500). Figure 14 shows that regardless of how the sulfur feed is varied (changing concentration vs. changing power generation), the yearly costs scale directly with sulfur removed.

2.a. Electrical

The pumps and compressors have been assumed to account for the majority of the electrical requirements for the HGD processes. The additional power requirements for lighting and instrumentation have been assumed to be 20% of the compressor and pump requirements for the base case of each HGD. It is assumed that the additional power requirements will not vary significantly with plant size.

The DSRP power requirement is significantly higher than that of the AHGP. The PRESAIR air compressor is the reason for the high DSRP power requirement. The air compressor supplies air to the regenerator for the oxidation of sulfurized sorbent. It is interesting to note that the cost of supplying oxygen by compressing air is more than the cost of separating oxygen and then compressing the pure oxygen. The phenomenon is not unprecedented; it has

been observed that as the pressure of injection is raised the cost of compressing air increases faster than the cost of separating oxygen and pressuring only oxygen (Hvizdos).

The compressed nitrogen feed to the DSRP - based process regenerator that is included in the air stream will increase the total volumetric flow to the turbine. This would indicate that there should be a power credit associated with the nitrogen's introduction, offsetting some of the compression costs. However, nitrogen will also increase the heat capacity of the stream, lowering the combustion temperature, thus lowering the power production. These competing effects have been assumed to cancel each other out. The design work assumes there is no change in power production attributed to the introduction of nitrogen.

2.b. Cooling Water

The steam generation/cooling loop is closed; maintaining water purity is not difficult for a self-contained loop. Furthermore, makeup water requirements will be negligible, for the detail level of this report. The is no debit calculated for the HGD steam system water because of the above mentioned reasons.

The steam condensate is assumed to be cooled to 90°C by cooling tower water. Tower water is exposed to the atmosphere, which means maintaining water purity will be an issue. There will also be makeup water requirements. Therefore a yearly debit has been calculated for the use of tower water. The tower water flow rates have been calculated in the Complete Steam Generation Scheme simulations (Appendix F). The tower water cools the steam stream that is considered "utilized." Utilized steam is a stream that was steam (441°C, 950 psia) but has been reduced to 30 psia and the corresponding bubble point temperature. Tower water cools the utilized steam stream to 90°C, before its reuse. The cost of the tower water is \$2.6x10⁻⁵/lb (Peters). The cost of the tower water is insignificant compared to the other yearly capital expenditures.

The cost of the tower is not an issue as there will already be a tower on site. HGD water sent to it will represent only a minor increase in load.

2.c. Oxygen

The cost of supplying oxygen has been assessed as a yearly expenditure with no capital cost. Dr. George Roberts indicated that its reasonable to expect oxygen to cost \$20/ton. The value is reasonable when compared with a dated guideline (Chilton, 1960) stating 99.5% pure oxygen at 450 psig would sell at \$8 to \$15/ton. There are no capital costs associated with the supplied oxygen assuming the oxygen will be bought from a gas supplier, in which only a usage charge is assigned. The price has been assumed to be set at \$20/ton, the price will actually be dependent on usage. The unit cost of oxygen decreases as quantity purchased increases.

There are oxygen costs only for the AHGP, since air is used to oxidize the sorbent in the DSRP.

2.d. Additional Employees

The number of additional employees required to operate the HGD processes have been assumed constant with process size. The additional employees required will depend more upon the complexity of the process than its size. The hiring of two additional engineers and two maintenance personal have been assigned to the DSRP. The AHGP has the hiring of three engineers accounted for. An additional engineer is hired since the AHGP is a more complex process to control because SO₂ production and consumption must be balanced. Furthermore, the purity of the SO₂ loop must be maintained. Two maintenance personnel are also accounted for in AHGP costs. The unit cost for an engineer is assumed to be \$100,000/year, and maintenance personnel are assumed to cost \$70,000/year. These numbers include the base salary and benefits.

2.e. Consumed Coal Gas

Coal gas (H₂ and CO) is consumed in both HGD processes. The consumption reduces the amount power that can be produced. The cost of consumed coal gas is calculated from the CO and H₂ lost during HGD, and calculating the value of the energy that the CO and H₂ could have produced. Calculation of power generation is described in Appendix J.

The DSRP consumes substantially more coal gas then the AHGP; this is the major factor in the lower yearly operating cost of the AHGP.

2.f. Additional Yearly Expenditures

Sorbent and catalyst attrition have not been accounted for in this report. The rate at which sorbent and catalyst need to be replaced times their unit cost will represent another yearly expenditure. Assuming the attrition costs for both processes are identical a comparison of the process economics will be unaffected by the absence of attrition costs in this report.

Maintenance charges have not be fully accounted for in this report. While the cost of additional employees to maintain equipment has been included, the cost of the replacement parts and equipment have not. Yearly maintenance costs should increase with years of service as well as with the size of the HGD process.

3. Economic Summary

The AHGP has a higher initial startup costs, indicated by its larger capital requirements. However, the AHGP has lower yearly expenditures then the DSRP. The operating cost difference is large enough to offset the initial startup cost difference within a few years.

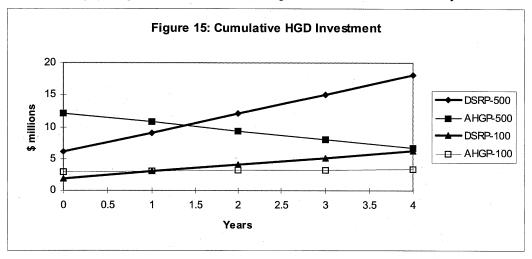


Figure 15 shows that despite an higher initial investment, within two years the AHGP can financially outperform the DSRP.

VIII. SUMMARY

Mass and energy balances were calculated for the Direct Sulfur Recovery Process - based Hot Gas Desulfurization and the Advanced Hot Gas Process. Establishing the balances has helped determine the equipment requirements for both processes. The specifications for the major pieces of equipment have been described in this report.

Simulating the HGD processes revealed the complexity of both processes. The AHGP appears to be the more difficult of the two processes to operate. More employees may be needed to operate the AHGP process than the DSRP -based process.

Capital costs for the AHGP are higher than those for the DSRP. However, yearly operating costs for the AHGP are considerably less than those of the DSRP. After two years of operation the total cost of implementing an AHGP will be less then the cost of a DSRP -based process. It will be more difficult to operate an AHGP but the substantial savings the process delivers makes it the more desirable process to implement.

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Appendix A Calculation of the SO₂ Circulation Rate for AHGP

 SO_2 circulation rates are set to create the desired flow conditions in the regenerating reactor. First the sorbent flow rate through the regenerator must be determined. Al_2O_3 must pass through the reactor in large quantities to keep the adiabatic temperature raise small. The sorbent flow is used to determine the reactor's cross sectional area. The SO_2 circulation rate necessary to provide a 2.5 cm/s upwards velocity is then calculated. Calculation results follow:

SO2 Regenerator Sizing - Co	AGHP	abodiment AHGP-b (SO2 Regen)		AHGP-c	
Givens:	Case E-2	Case E-2			
Sorbent circulation rate, lb/hr	166010		496000		48000
Sorbent bulk density, lb/ft3	62.4		62.4		62.4
Reg'd rxtr residence time, hr	1		1		1
Regen Gas v _{super} , cm/sec	2.5	r en	2.5		2.5
Desired H/D	. 2		2		2
Adjusted values:					
Assumed Bed Depth, ft	10		10		10
SO2 needed ft3/hr	79,813		238,462		23,077
Calculated values:	•				
Hold-up volume, ft3	2660	ı	7949		769
Diameter, ft	18		32		9.9
X-section area, ft2	266	· •	795		77
Calculated H/D	0.54		0.31		1.01
RG Vol. flow rate, acf/sec	21.8		65.2		6.3
RG flow rate, lb/hr	86366		258043		24972
Ratio of RG flow/sorbent, lb/lb	0.52		0.52		0.52
Calculated Bed Depth, ft					
Operating conditions/Gas Density	Calc'ns:				
Pressure, psig	275		275		275
Pressure, psia	289.7	•	289.7		289.7
MW of gas	64		64		64
Bed Temp., C	600		600		600
Bed Temp., R	1571.67		1571.67		1571.67
R, gas constant,	10.73		10.73		10.73
Gas density, lb/ft3	1.1		1.1		1.1

Appendix B Heat Transfer Coefficients

The following approximate overall heat transfer coefficients were found in the literature. The values were in used estimating the heat exchangers' overall heat transfer coefficients.

	Coolers	
Hot Fluid	Cold Fluid	Overall U _D , BTU/hr ft ² °F
Water	Water	250 - 500
Gases	Water	2 - 50
	Heaters	
Hot Fluid	Cold Fluid	Overall U _D , BTU/hr ft ² °F
Steam	Water	200 - 700
Steam	Gases	5 - 50

Values above found in Kern (1950).

Fluid combination	U, BTU/hr ft ² °F
Water to compressed air	10 - 30
Water to water	150-275
Steam to aqueous solutions	100-600
Steam to gases	5 - 50

Values above found in Welty, Wicks, and Wilson (1984).

Appendix C **Determination of Catalyst Velocity in DSRP Reactor**

In order to determine whether the catalyst in the DSRP Reactor (a fast fluid-bed reactor) will be transported to the top of the reactor by the gas feed, the following calculation was performed. A terminal velocity calculation was performed on a catalyst particle. This calculation will approximate the catalyst's velocity relative to the gas phase. The gas velocity through the DSRP will be 3 ft/s (0.9 m/s). The catalyst's relative velocity needs to be less than the gas velocity in order for the catalyst to be elevated.

Terminal velocity is determined from a force balance on the particle.

$$m\frac{dv}{dt} = F_g - F_d - F_b$$

$$m\frac{dv}{dt} = mg - \frac{C_D v^2 \rho (\pi D_p^2 / 4)}{2} - \frac{m\rho g}{\rho_p}$$

At steady state the left side equals zero and the equations simplify to give the steady state (terminal) velocity:

$$v_{ss} = \sqrt{\frac{4}{3}(\frac{D_pg}{C_D\rho})(\rho_p - \rho)}$$

The catalyst size is 160 micron.

$$D_p = 1.6 \times 10^{-4} \text{ m}$$

$$\rho_p = 1.2 \text{ g/cm}^3$$
 $g = 9.8 \text{ m/s}^2$

$$g = 9.8 \text{ m/s}^2$$

Bulk samples of the catalyst have a density (ρ_{bulk}) of 0.9 g/cm³. The bulk catalyst is assumed to have a packing fraction of 0.74, the highest packing fraction possible for spheres. Assuming the packing fraction enables calculation of the individual catalyst density (ρ_p) .

$$\rho_p = \rho_{bulk} / (packing fraction)$$

$$\rho_{\rm p} = 1.2 \text{ g/cm}^3 = (0.9 \text{ g/cm}^3) / (0.74)$$

The gas density is taken as a weighted average of the feeds ROG-COOL and SLIPSTRM.

$$\rho = 0.50 \text{ lb/ft}^3 \text{ x } (1,000 \text{ gr}) / (2.205 \text{ lb}) \text{ x } (1 \text{ ft}^3) / (30.48 \text{ cm})^3 = 0.008 \text{ g/cm}^3 = 8 \text{ kg/m}^3$$

Inserting the values gives:

$$v_{ss} = \sqrt{\frac{0.3136 \frac{m^2}{s^2}}{C_D}}$$

The drag coefficient C_D is correlated with the Reynolds number (N_{Re}) of the gas phase. After determining the Reynolds number C_D can be determined from charts in Bird (1960).

$$N_{Re} = D_p v_{ss} \rho / \mu$$

The steady state velocity is determined iteratively. That leaves μ , viscosity of the gas, the only other unknown.

For viscosity calculations, the gas will been assumed to have the properties of nitrogen (N₂ represents over 50 wt% of reactor gas).

Reactor conditions $T = 600^{\circ}C$ and P = 275 psia.

The Reichenberg correlation was used for the determination of the high pressure viscosity (Perry's 3-279). The correlation typically has errors of less than 10 percent.

Equations

$$(\mu - \mu^{o})/(\mu^{o} P) = A P_{r}^{1.5} / [B P_{r} + (1 + C P_{r}^{D})^{-1}]$$

$$A = 1.9824 \times 10^{-3} T_{r}^{-1} \exp (5.2683 T_{r}^{-0.5767})$$

$$B = A (1.6552 T_{r} - 1.2760)$$

$$C = 0.1319 T_{r}^{-1} \exp (3.7035 T_{r}^{-79.8678})$$

$$D = 2.9496 T_{r}^{-1} \exp (2.9190 T_{r}^{-16.6169})$$

Nitrogen Properties

$$T_r = T \ / \ T_c = 873 \ K \ / \ 126.2 \ K = 6.91 \qquad P_r = P \ / \ P_c = 275 \ psia \ / \ 492 \ psia = 0.559 \\ \mu^o = \mu \ (1 \ atm, \ 873 \ K) = 3.8 \ x \ 10^8 \ Poise \\ And \ for \ nonpolar \ molecules: \ P = 1$$

$$A = 0.001615 \quad B = 0.0164 \quad C = 0.01909 \quad D = 0.4269$$

$$(\mu - \mu^{\circ}) / (\mu^{\circ}) = 6.7498 \times 10^{-4} / 0.9945 = 6.787 \times 10^{-4}$$

$$\mu = 3.8 \times 10^{-4} + (3.8 \times 10^{-4})(6.787 \times 10^{-4}) \text{ Poise}$$

$$\mu = 3.8 \times 10^{-4} \text{ Poise} = 3.8 \times 10^{-5} \text{ Pa s} = 3.8 \times 10^{-5} \text{ kg/(m s)}$$

The Reynolds number is can now be expressed:

$$N_{Re} = (1.6 \text{ x } 10^{-4} \text{ m}) (v_{ss}) (8 \text{ kg/m}^3) / [3.8 \text{ x } 10^{-5} \text{ kg/(m s)}]$$

 $N_{Re} = (v_{ss}) 33.68 \text{ s/m}$

And our velocity equation is:
$$v_{ss} = \sqrt{\frac{0.3136 \frac{m^2}{s^2}}{C_D}}$$

First Iteration, take $v_{ss}=0.9$ m/s then $N_{Re}=30$ (above equation) For the above Reynolds number $C_D=2.4$ (Fig. 6.3-1 in Bird) Velocity equation gives $v_{ss}=0.36$ m/s

The calculations are repeated.

Second iteration:

$$v_{ss} = 0.36 \text{ m/s}$$

$$N_{Re} = 12$$

$$C_D = 4.2$$

$$v_{ss} = 0.27 \text{ m/s}$$

Third iteration:

$$v_{ss} = 0.27 \text{ m/s}$$

$$N_{Re} = 9.09$$

$$C_D = 4.9$$

$$v_{ss} = 0.252 \text{ m/s}$$

Fourth iteration:

$$v_{ss}$$
 0.252 m/s

$$N_{Re} = 8.49$$

$$C_D = 5.13$$

$$v_{ss} = 0.247 \text{ m/s}$$

Fifth iteration:

$$v_{ss} 0.247 \text{ m/s}$$

$$N_{Re} = 8.33$$

$$C_D = 5.19$$

$$v_{ss} = 0.246 \text{ m/s}$$

The velocity of falling catalyst is 0.25 m/s. Thus in a gas stream flowing up at 0.9 m/s the catalyst will rise at 0.65 m/s (2.1 ft/s).

CONCLUSION: The gas stream will be capable of elevating the catalyst.

Sorbent in the risers will be elevated at approximately the same velocity (20 ft/s) as the gas lifting it.

Appendix D Calculation of DSRP Catalyst Cycling Rate

The rate at with catalyst is fed to the DSRP was determined by the heat removal requirements of the DSRP reactor. Heat is removed from the reactor by cooling the catalyst effluent and reintroducing that catalyst. Exiting catalyst temperature is set at 600°C and the catalyst is cooled to 500°C.

Catalyst Properties

The DSRP reactor catalyst is a porous aluminum oxide catalyst modeled as Al_2O_3 . Catalyst density at ambient conditions is 56.18 lb/ft³. This density includes the void space filled by air. ASPEN was utilized to determine the void space in the settled catalyst, assuming nitrogen fills the voids in the solid catalyst. At ambient conditions 1 lb of Al_2O_3 and 0.00095 lb of N_2 have a combined density of 55.6 lb/ft³.

The similar densities allow us to assume that there is roughly 0.00095 lb of nitrogen present for every 1 lb of solid Al_2O_3 (at ambient conditions). That quantity of nitrogen occupies 0.0132 ft³ (at ambient conditions). This represents the catalyst void volume and is expected to remain constant.

$$V_{\text{void}} = 0.0132 \text{ ft}^3 / 1 \text{ lb Al}_2\text{O}_3$$

The density of the gas in the reactor was taken as the average of nitrogen's density at 600°C (275 psia) and 500°C (275 psia).

$$\rho_{\rm gas} = 0.483 \, \text{lb/ft}^3$$

Therefore the mass of gas (in the settled catalyst) per pound Al₂O₃ can be calculated.

$$M_{gas} = 0.483 \text{ lb/ft}^3 \times 0.0132 \text{ ft}^3 = 0.0064 \text{ lbs}$$

The heat transfer requirements for cooling Al_2O_3 were than simulated (including cooling nitrogen contained in the catalyst voids).

$$Q(600^{\circ}C -> 500^{\circ}C) = -51.239 \text{ BTU/lb Al}_{2}O_{3}$$

Calculation of necessary catalyst circulation rate:

(circulation rate {lb/hr}) =
$$(Q_{DSRP}) / (-51.239 BTU/lb Al_2O_3)$$

- **DSRP** (circulation rate {lb/hr}) = $(-15,340,000 \text{ BTU/hr}) / (-51.239 \text{ BTU/lb Al}_2\text{O}_3)$ = 300,000 lb Al₂O₃ / hr
- **DSRP-b** (circulation rate {lb/hr}) = $(-51,320,000 \text{ BTU/hr}) / (-51.239 \text{ BTU/lb Al}_2\text{O}_3)$ = $1,000,000 \text{ lb Al}_2\text{O}_3 / \text{hr}$
- **DSRP-c** (circulation rate {lb/hr}) = $(-4,029,000 \text{ BTU/hr}) / (-51.239 \text{ BTU/lb Al}_2\text{O}_3)$ = **79,000 lb Al}_O_3 / hr**
- **DSRP-100** (circulation rate {lb/hr}) = $(-6,459,000 \text{ BTU/hr}) / (-51.239 \text{ BTU/lb Al}_2\text{O}_3)$ = **130,000 lb Al}_03 / hr**
- **DSRP-500** (circulation rate {lb/hr}) = $(-31,370,000 \text{ BTU/hr}) / (-51.239 \text{ BTU/lb Al}_2O_3)$ = **610,000 lb Al}_O3 / hr**

Appendix E Process Flowsheets and Stream Summaries

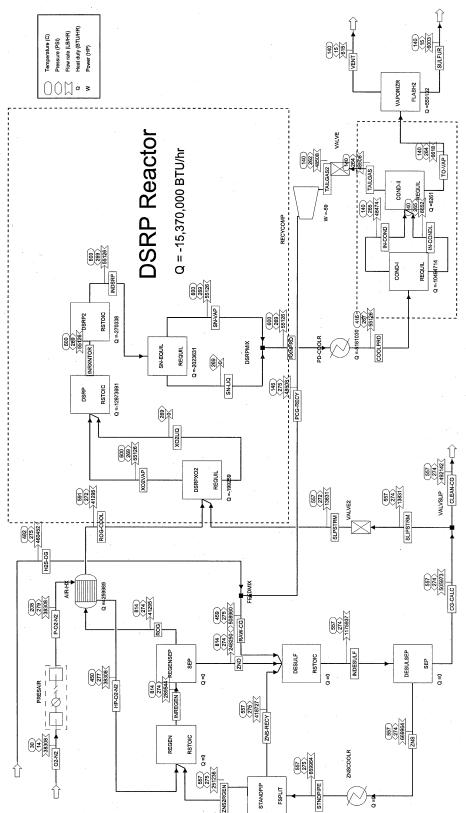
Direct Sulfur Recovery Process Simulations

DSRP (base case)	0.85 mole% H ₂ S	260 MW generated
DSRP-b	2.50 mole% H ₂ S	260 MW generated
DSRP-c	0.25 mole% H_2S	260 MW generated
DSRP-100*	0.85 mole% H ₂ S	110 MW generated
DSRP-500	0.85 mole% H_2S	540 MW generated

Advanced Hot Gas Process Simulations

AHGP (base case)	$0.85 \text{ mole}\% \text{ H}_2\text{S}$	260 MW generated
AHGP-b	2.50 mole% H ₂ S	260 MW generated
AHGP-c	0.25 mole% H ₂ S	260 MW generated
AHGP-100*	$0.85 \text{ mole}\% \text{ H}_2\text{S}$	110 MW generated
AHGP-500*	0.85 mole% H ₂ S	540 MW generated

^{*}DSRP-100, AHGP-100, and AHGP-500 were not simulated. The flowrates and heat duties will scale directly from the base cases (DSRP and AHGP). DSRP-100 and AHGP-100 values equal DSRP and AHGP values scaled by 0.4211. AHGP-500 values equal AHGP values scaled by 2.1055.



High Pressure Condenser Q = -10,500,000 BTU/hr

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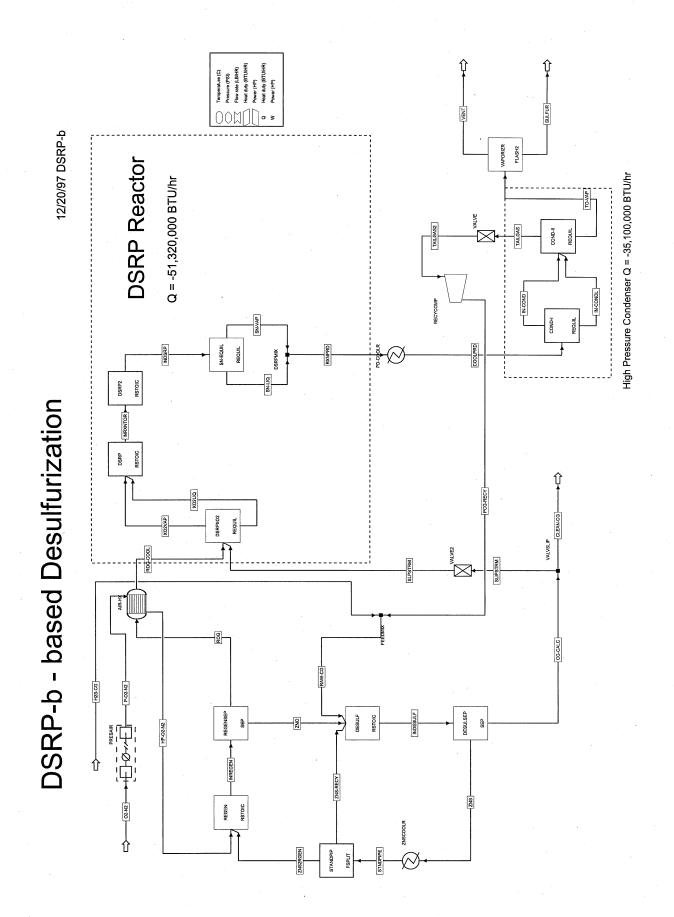
isplay ALLS	TREAMS	CG-CALC	CLEAN-CG	COOLPRD		2		N-CONDL	INDESULF	NDSRP	INREGEN	INRXNTOR		Display ALLS	TREAMS
nits:	From	_		PD-COOLR			COND-I	COND-I	DESULF	SRP2	REGEN			Units: From	From
ormat: SOLI	DS To	VALVSLIP		COND-I		REGEN		COND-II	DESULSEP	SN-EQUIL	REGENSEP DSRP2			Format: SOLI	DS To
	Phas	VAPOR	VAPOR	VAPOR				LIQUID	VAPOR	/APOR	VAPOR				Phas
emperature	<u></u>	557.2	557.2	415	482.2	450	140	140		009	814.4	009	30	30 Temperature	<u>[</u>
ressure	[PSI]	274.4		266.6	275	276.9	264.6	264.6	274.4	268.6	273.6	268.6		13.7 Pressure	[PSI]
lass VFrac		1	1	1	1	1	1	0	0.43	1	0.143	1	1	Mass VFrac	
lass SFrac		0	0	0	0	0	0	0	0.57	0	0.857	0	0	0 Mass SFrac	
** ALL PHAS	ES ***													*** ALL PHAS ES ***	ES ***
lass Flow	[LB/HR]	505972.781	492141		460451.688	38308.016	48474.25	6651.654	1.18E+06	55125.902	١,,		38308.016 Mass Flow	Mass Flow	[LB/HR]
Volume Flow	[CUFT/HR]	1.37E+06	1.33E+06	87129.523	1.15E+06	67432.328	49166.316	48.953	1.37E+06	112094.914	96496.891	111308.695	567372.75	567372.75 Volume Flow	[CUFT/HR]
nthalpy	[BTU/HR]	-1.15E+09	-1.11E+09	-6.57E+07	-1.09E+09	7.31E+06	-7.15E+07	-4.68E+06	-4.84E+09	-5.85E+07	-1.38E+09	-5.82E+07	79241.141 Enthalpy	Enthalpy	[BTU/HR]
ensity	[LB/CUFT]	0.37		0.633	0.399	0.568	986'0	135.877	0.857	0.492	3.001	0.495	0.068	0.068 Density	[LB/CUFT]
lass Flow	[LB/HR]													Mass Flow	[LB/HR]
2S		63.71	61	63.879			2.69	0.18	63.71	63.879	11909.276	1787.733		028	
125		6.82	9	72.073	6270.481		71.958	0.116	6.82	72.072		72.072		HZS	
20		78082.43	75947.969		70525.25		4175.116	788.209	78082.43	4963.325		4963.325		HZO	
2		< 0.001	: 0.001	2631.176			< 0.001	0.212	0.212 < 0.001	5861.615		3273.283		S2	
9		0.39	0.38				0.965	1890.29	0.39	0.011		0.011		Se Se	
80		2.764	2	1339.271			2.025	3968.209	2.764	0.076		0.076		8S	
o.		218164.266	212200.516	2.262	218162		2.262	2.262 < 0.001	218164.266	2.262		2.262		CO	
:02		130332.672	126769.898	12929.425	117407.195		12925.243	4.181	130332.672	12929.425		10560.996		CO2	
2		11766.221	11444.579	0.85	11765.37		0.85	0.85 < 0.001	11766.221	0.85		0.85		H2	
22						8922.588							8922.588 02	02	
2		67553.203	65706.57	31232.063	36321.383	29385.428	31231.807	0.256	67553.203	31232.063	29385.428	31232.063	29385.428 NZ	N2	
SOS		0.323	0.314	0.323			0.323	0.323 < 0.001	0.323	0.323		3233.392		SOO	
NO											15129.819			SNO	
NS									144457.359		36055			SNZ	
L203									525506.5		197064.938		-	AL203	

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		_			_	_	_		_			_	_		_	_	_	_										
TREAMS	From	DS To	Phas	<u></u>	[PSI]			ES ***	[LB/HR]	[CUFT/HR]	[BTU/HR]	[LB/CUFT]	[LB/HR]															
Display ALLS	Units: From	Format: SOLI DS To		140 Temperature	14.7 Pressure	0 Mass VFrac	0 Mass SFrac	*** ALL PHAS ES ***	6002.956 Mass Flow	33.124 Volume Flow	Enthalpy	Density	Mass Flow	028	H2S	H20	S2	Se	S8	8	C02	H2	02	NZ	SOO	SNO	SNZ	AL203
1			LIQUID			0	0			33.124	-4.57E+05 Enthalpy	181.225 Density		0.001 028	< 0.001	146.193 H2O	0.212 S2	724.696 S6	5131.851 S8	trace CO	0.002	trace		trace	trace			
STNDPIPE		STANDPIP	MISSING	557.2	275	0	1		669963.875	3189.083	-3.69E+09	210.08															144457.359	525506.5
SN-VAP				009	268.6	1	0		55125.902	109933.273	-6.05E+07	0.501		63.879	72.072	4963.325	2631.176	1891.255	1339.271	2.262	12929.425	0.85		31232.063	0.323			
SN-LIQ		DSRPMIX	MISSING						0		0.00E+00																	
SLPSTRM		DSRPX02	VAPOR	557.2	271.6	1	0		13831.278	37813.297	-31319000	0.366		.1.742	0.186	2134.462	trace	0.011	0.076	5963.741	3562.775	321.642		1846.635	600'0			
SLIPSTRM		VALVE2	VAPOR	557.2	274.4	1	0		13831.278	37429.125	-3.13E+07	0.37		1.742	0.186	2134.462	trace	0.011	0.076	5963.741	3562.775	321.642		1846.635	600:0			
ı	DSRPMIX	PD-COOLR VALVE2	VAPOR	009	268.6	1	0		55125.902	109933.391	-6.05E+07	0.501		63.879	72.073	4963.325	2631.176 trace	1891.255	1339.271	2.262	12929.425	0.85		31232.063	0.323			
ROG-COOL RXNPRD		2	VAPOR	2005	271.6	1	0		41294.703	76325.195	-1.38E+07	0.541		11909.276										29385.428				
ROG	REGENSEP AIR-HX	AIR-HX	VAPOR	814.4	273.6	-	0		41294.703	95300.984	-9.58E+06	0.433		11909.276										29385.428				
l			VAPOR	458.8	275	-	0		508959.5	1.20E+06	-1.16E+09	0.423		63.71	6342.445	74733.492	< 0.001	0.39	2.764	218164.266	130332.672	11766.221		67553.203	0.323			
PCG-RECY RAW-CG	RECYCOMP FEEDMIX	FEEDMIX		146.5	275	1	0		48507.805	48127.191	-7.15E+07	1.008		63.71	71.964	4208.244	< 0.001	0.39	2.764	2.262	12925.476	0.85		31231.822	0.323			
P-02-N2	PRESAIR	AIR-HX	VAPOR	207.6	278.9	٦	0		38308.016	44510.953	3.05E+06	0.861											8922.588	29385.428				-
TREAMS	From	DS To	Phas	<u></u>	[PSI]			ES ***	[LB/HR]	[CUFT/HR]	[BTU/HR]	(LB/CUFT)	[LB/HR]															
Display ALLS TREAMS P-02-N2	Units:	Format: SOLI DS To		Temperature	Pressure	Mass VFrac	Mass SFrac	*** ALL PHAS ES ***	Mass Flow	Volume Flow	Enthalpy	Density	Mass Flow	028	HZS	HZO	S2	Se	S8	8	C02	H2	02	NZ	SOO	ONZ	SNZ	AL203

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AL203	197064.938	328441.563	525506.5	197064.938							
SNZ	54171.508	90285.852	144457.359	36055							
ONZ				15129.819				-			
cos					600.0		< 0.001	< 0.001		0.323	
N2					31232.063		0.241	1.822 0.241	3123	31231.822	
02											
H2					425.192		< 0.001	0.85 < 0.001		0.85	
CO2					5823.677		3.948	3.95	1292	12925.476	
00					4524.921		< 0.001	2.262 < 0.001		2.262	
88					0.076		1.567	5133.418		2.764	
98					0.011		0.221	724.917	66.0	68:0	
S2					trace		0.212 < 0.001		< 0.001	< 0.001	
H2O					1209.115		608.888	755.081	4208.244	4208.244	
H2S					0.186				71	71.964	
028					11911.018	-	0.169	0.17	9	63.71	
Mass Flow [LB/HR]											[LB/HR]
Density [LB/CUFT]	210.08	210.08	210.08	207.583	0.466		0.034	142.868	926'0	0.985	LB/CUFT]
y	-1.38E+09	-3.69E+09 -2307300000	-3.69E+09	-1	-4.5	0.00E+00	-3.4	4	-7.17E	-7.17E+07	(BTU/HR)
Volume Flow [CUFT/HR]	1195.906	1993.177	3189.083		118222.719		18288.283	46.323	49703.582	49255.441	[CUFT/HR]
Mass Flow [LB/HR]	251236.453	418727.406	669963.875	248249.766	55126.27	0	615.143	6618.099	48507.805	48507.805	[LB/HR]
*** ALL PHAS ES ***											ES ***
	1	1	1	1	0		0	0	0	0	
Mass VFrac	0	0	0	0	1		1	0	1	1	
	275	275	274.4	273.6	268.6		14.7	264.4		264.4	[bSI]
Temperature [C]	557.2	č			009		140		139.9	140	[0]
	MISSING		MISSING	MISSING	VAPOR	MISSING	VAPOR	LIQUID	VAPOR	VAPOR	has
Format: SOLI DS To	REGEN		DESULF ZNSCOOLR DESULF	DESULF		DSRP		VAPORIZR	RECYCOMP	VALVE	DS To
	STANDPIP		DESULSEP	REGENSEP	20	DSRPX02	VAPORIZR	COND-II	VALVE		шо
	ZNSZRGEN		SNZ	ZNO		XOZLIQ	VENT				TREAMS



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															_			_	_	_	_						_	_
TREAMS	From	DS To	Phas	[0]	[PSI]			ES ***	[LB/HR]	[CUFT/HR]	[BTU/HR]	[LB/CUFT]	[LB/HR]															-
Display ALLS	Units:	Format: SOLI		30 Temperature	13.7 Pressure	Mass VFrac	0 Mass SFrac	*** ALL PHAS	121500 Mass Flow	1799510 Volume Flow	Enthalpy	0.068 Density	Mass Flow	028	42S	HZO	32	98	8S	00	CO2	42	22	72	SOO	ONZ	SNZ	4L203
O2-N2		PRESAIR		30	13.7	1	0		121500	1799510	2.51E+05 Enthalpy	890.0											28299.416 02	93200.586 NZ				
INRXNTOR	DSRP	DSRP2 F	VAPOR	009	267.6	1	0		184607.719	376264.813	-1.99E+08	0.491		5444.741	240.342	17155.324	10711.036	0.105	0.734	68.9	36256.879	2.783		104942.266	9846.616	٠		
	REGEN	REGENSEP		810	272.6	660'0	0.901		1322550	307882.375	-6.55E+09	4.296		37772.176										93200.586		47986.641	202017.281	941570.188
		SN-EQUIL	VAPOR	900	267.6	1	0		184607.219	378670.75	-2.00E+08	0.488		195.105	240.342	17155.324	18593.258	0.105	0.734	68.9	43469.434	2.783		104942.266	0.985			
		DESULSEP	VAPOR	647.5	273.6	0.197	0.803		3324180	1900580	-1.60E+10	1.749		194.606	62.453	99935.453	0.695 < 0.001	1.28	8.967	231203.391	167878.578	12471.098		143432.969	0.984		576614.875	2092380
IN-CONDL	COND-I	COND-II	LIQUID	140	263.6	0	0		21287.002	159.053	-1.61E+07	133.836		0.526	0.371	2684.464	0.695	5882.623	12704.021	6.89 < 0.001	13.482	2.783 < 0.001		0.818	0.984 < 0.001			
		COND-II		140	263.6	1	0		163323.219	166528.75	-2.43E+08	0.981		194.579	239.971	14470.859	< 0.001	3.086	6.665	68.9	43455.953	2.783		104941.445	0.984			
HP-02-N2		REGEN		450	277.4	1	0		121500	213489.484	2.32E+07	0.569											28299.416	93200.586				
HZS-CG		FEEDMIX		482.2	275	1	0		501234.031	1244460	-1.16E+09	0.403			19916.91	74738.914				231196.5	124421.914	12468.315		38491.473				
	PD-COOLR	COND-I		415	265.6	1	0		184610.219	294846.875	-2.24E+08	0.626		195.105	240.342	17155.324	8658.119	5885.709	4053.262	68.9	43469.434	2.783		104942.266	0.985			
CLEAN-CG				647.5	273.6	1	0		601554.75	1733330	-1.24E+09	0.347		178.675	57.341	91754.563	< 0.001	1.175	8.233	212276.672	154135.734	11450.191		131691.281	0.904			
		VALVSLIP	VAPOR	647.5	273.6	-	0		655189.75	1887870	-1.36E+09	0.347		194.606	62.453	99935.453	< 0.001	1.28	8.967	231203.391	167878.578	12471.098		143432.969	0.984			
Display ALLS TREAMS CG-CALC	From	DS To	Phas	<u></u>	[PSI]			ES ***	[LB/HR]	[CUFT/HR]	[BTU/HR]	ILB/CUFT	[LB/HR]															
Display ALLS	Units:	Format: SOLI DS To		Temperature	Pressure	Mass VFrac	Mass SFrac	*** ALL PHAS	Mass Flow	Volume Flow	Enthalpy	Density	Mass Flow	028	H2S	HZO	S2	98	S8	00	C02	H2	02	N2	SOO	ONZ	SNZ	AL203

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						_		_					٠															
TREAMS	From	DS To		[C]	[PSI]			ES ***	[LB/HR]	[CUFT/HR]	[BTU/HR]	[LB/CUFT]	[LB/HR]															
Display ALLS	Units: From	Format: SOLI		140 Temperature	14.7 Pressure	0 Mass VFrac	0 Mass SFrac	*** ALL PHAS ES ***	Mass Flow	105.13 Volume Flow	Enthalpy	Density	Mass Flow	02S	H2S	420	32	36	38	8	202	H2	02	N2	SOO	ZNO	SNZ	AL203
				140	14.7 F	lo l	0	*	19044.756 Mass Flow	105.13	-1.45E+06 Enthalpy	181.154 Density	_	0.002 028	0.001 H2S	464.116 H2O	0.695 S2	2321.039 S6	16258.897 S8	trace	0.006 CO2	trace		< 0.001	trace	Ž	Z	/
STNDPIPE	ZNSCOOLR VAPORIZR	STANDPIP	MISSING	640	275	0	1		2668990	12704.87	-1.46E+10	210.076										-			•		576614.875	2092380
		DSRPMIX		009	267.6	_	0		184610.219	372001.031	-2.06E+08	0.496		195.105	240.342	17155.324	8658.119	5885.709	4053.262	68.9	43469.434	2.783		104942.266	0.985			
		DSRPMIX	MISSING						0		0																	
SLPSTRM	VALVSLIP VALVE2 SN-EQUIL	DSRPX02	VAPOR	647.5	270.6	-	0			ľ	-1.11E+08	0.343		15.931	5.113	8180.895	< 0.001		0.734		1	1020.906		11741.678	0.081			
SLIPSTRM	VALVSLIP	VALVE2	VAPOR	647.5	273.6	1	0	П	53635.004	154544.391	-1.11E+08	0.347		15.931	5.113	8180.895	< 0.001	0.105	0.734		-	1020.906		11741.678	0.081			
				009	267.6	1	0		184610.219	372001.438	-2.06E+08	0.496		195.105	240.342	17155.324	8658.119 < 0.001	5885.709	4053.262	68.9	43469.434	2.783		104942.266	0.985			
ROG-COOL	AIR-HX DSRPMIX	DSRPX02	VAPOR	586.3	270.6	1	0		130972.758	302165.188 241725.797 372001.438 154544.391	-4.42E+07	0.542		37772.176										93200.586				
ROG	SEP	AIR-HX		810	272.6	1	0	1 1	130972.758		-3.07E+07	0.433		37772.176										93200.586				
RAW-CG	FEEDMIX	DESULF	VAPOR	418.7	275	-	0		664662.5		-1.40E+09	0.471		194.606	L	89313.766	< 0.001	1.28	8.967	1	1	12471.098		143432.969	0.984			
PCG-RECY		FEEDMIX	VAPOR	147.5	275	1	0		163428.5	٢	-2.43E+08	1.004		194.606	239.99	14574.853	< 0.001	1.28	8.967	68.9	43456.652	2.783		104941.492	0.984			
	PRESAIR			.7	279.4	1	0		121500	140938.547	9691570	0.862											28299.416	93200.586				
TREAMS	From	DS To	Phas		[PSI]			ES ***	[LB/HR]	[CUFT/HR]	[BTU/HR]	[LB/CUFT]	[LB/HR]															
Display ALLS	Units:	Format: SOLI		Temperature	Pressure	Mass VFrac	Mass SFrac	*** ALL PHAS	Mass Flow	Volume Flow	Enthalpy	Density	Mass Flow	028	HZS	HZO	S2	98	88	00	CO2	Н2	02	N2	cos	SNO	SNZ	AL203

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													_		_	_	_											
TREAMS	From	DS To	Phas	<u> </u>	[PSI]			S ES ***	[LB/HR]	[CUFT/HR]	[BTU/HR]	[LB/CUFT]	[LB/HR]															
Display ALLS	Units:	Format: SOLI		Temperature	Pressure	Mass VFrac	Mass SFrac	*** ALL PHAS	Mass Flow	Volume Flow	Enthalpy	Density	Mass Flow	028	HZS	HZO	S2	Se	88	8	C02	H2	02	NZ	SOS	ONZ	SNZ	AL203
ZNSZRGEN	STANDPIP	REGEN	MISSING	640	275	0	-		1201050	5717.191	-6.57E+09	210.076															259476.688	941570.188
	STANDPIP		MISSING	640	275	0	-		1467950	829.678	-8.03E+09	210.076															317138.188	1150810
SNZ	DESULSEP	ZNSCOOLR	MISSING	647.5	273.6	0	1		2668990	12704.87	-1.46E+10	210.076															22	2092380
ZNO	REGENSEP		MISSING	810	272.6	0	1		1191570	5717.191	-6.52E+09	208.419														47986.641	202017.281	941570.188
	DSRPX02		VAPOR	009	267.6	1	0		184608.375	398196.594	-1.59E+08	0.464		37788.105	5.113	4870.572	< 0.001	0.105	0.734	13779.622	21830.447	1391.339		104942.266	0.081			
	DSRPX02	DSRP	MISSING						0		0																	
	VAPORIZR D	Ц	VAPOR N	140	14.7	-	0		2136.962	63551.742	-1.21E+07	0.034		0.497		2116.355		0.776	5.434	< 0.001	12.774	< 0.001		0.773	< 0.001			
		VAPORIZR	LIQUID	140	263.6	0	0		21181.719	150.956	-1.54E+07	140.317		0.499	0.353	2580.471	0.695 < 0.001	2321.814	16264.332	< 0.001	12.781	< 0.001		0.773	< 0.001			
TAILGAS2	VALVE	RECYCOMP		_	261	1	0		163428.5	168340.75	-2.43E+08	0.971		194.606	239.99	14574.853	< 0.001	1.28	8.967	68.9	43456.652	2.783		104941.492	0.984			
TAILGAS	COND-II		VAPOR	140	263.6	ľ	0		163428.5	166691.297	-2.43E+08	86'0		194.606	239.99	14574.853	< 0.001	1.28	8.967	68'9	43456.652	2.783		104941.492	0.984			
TREAMS			Phas	<u></u>	[PSI]			ES ***	[LB/HR]	[CUFT/HR]	[BTU/HR]	[LB/CUFT]	[LB/HR]															
Display ALLS	Units:	Format: SOLI		Temperature	Pressure	Mass VFrac	Mass SFrac	*** ALL PHAS	Mass Flow	Volume Flow	Enthalpy	Density	Mass Flow	028	H2S	HZO	S2	Se	S8	8	CO2	H2	02	N2	cos	ONZ	SNZ	AL203

