Control of Sulfur Emissions from Oil Shale Retorts

by

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#### **FOREWORD**

When energy and material resources are extracted, processed, converted, and used, the related impacts on our environment and even on our health often require that new and increasingly more efficient pollution control methods be used. The Industrial Environmental Research Laboratory — Cincinnati (IERL-Ci) assists in developing and demonstrating new and improved methodologies that will meet these needs.

Synthetic fuel processes under development must be characterized prior to commercialization so that pollution control needs can be identified and control methods can be integrated with process designs. Shale oil recovery processes are expected to have unique air, water, and solid waste pollution control requirements. This report describes an in-depth evaluation of the control technology systems that are applicable to the removal of hydrogen sulfide from retort off-gases. Further information on the environmental aspects of oil shale processing and control technology can be obtained from IERL-Ci, Oil Shale and Energy Mining Branch.

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#### ABSTRACT

The objectives of this study were to determine the most applicable control technology for control of sulfur emissions from oil shale processing facilities and then to develop a design for a mobile slipstream pilot plant that could be used to test and demonstrate that technology.

The work conducted included an in-depth evaluation of available gas characterization data from all major oil shale development operations in the United States. Data gaps and inconsistencies were identified and corrected where possible through working with the developers and/or researchers in the field. From the gas characterization data, duty requirements were defined for the sulfur removal systems. Based on this information, Stretford gas sweeting technology was recommended, and the design of a 1000 CFM pilot plant was completed.

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#### I. PROJECT SUMMARY

#### A. INTRODUCTION

The future beneficial use of the nation's extensive oil-shale resources depends not only on the development of suitable process economics but also on the development of suitable environmental controls. Even though the oil that would be produced is comparatively low in sulfur content, the potential sulfur emissions from large-scale production of shale oil could be enormous. Oil shale contains up to about 2% sulfur. A typical shale in the Green River Formation in Colorado contains about 0.7% sulfur. When the shale is retorted, somewhere between 16 and 30% of the sulfur is liberated to the gas stream, with the majority remaining with the spent shale. The emissions from a 64,000-m<sup>3</sup>/day cil-shale industry could be as high as 691-1273 tonnes per day if emission controls were not applied. If conventional flue-gas scrubbing systems were used to control the emissions for which the average reduction achieved is 90%, the controlled emissions would be 69 to 127 tonnes day. However, if the sulfur could be removed before the gas is burned, in which case the average reduction is 98%, the controlled emissions would be in the order of 14 - 26 tonnes/day.

Gases produced by direct-fired retorts, either above ground or in-situ, are significantly different from gases normally encountered in application of desulfurization technology that the technology cannot just be transferred. Gases from direct-fired retorts contain large amounts of inert components and have a high ratio of carbon dioxide  $(CO_2)$  to hydrogen sulfide  $(H_2S)$ ; they also contain large amounts of ammonia and unsaturated hydrocarbons, such as acetylene, ethylene, propylene, butylene, and butadiene. The gases are saturated with water and contain some oxygen and trace amounts of sulfur species other than  $H_2S$ .

The large amounts of  ${\rm CO}_2$  in the gases and the high  ${\rm CO}_2/{\rm H}_2{\rm S}$  ratios make it impractical to employ many of the desulfurization technologies. Since the gases are produced in huge volumes at near-atmospheric pressures, many other desulfurization processes cannot be economically

applied. Those processes that can be applied may be only marginal in performance because of the large amounts of  ${\rm CO}_2$  and the presence of oxygen and/or unsaturated hydrocarbons in the gases or because the gas may contain a large amount of organic sulfur.

Oil-shale developers are involved in a number of significant pilot-scale activities for the development of retorting process technology, and have indicated to the United States Environmental Protection Agency (EPA) their willingness to cooperate on joint projects for sulfur control technology evaluation. To capitalize on this opportunity and to explore the possibility that sulfur emission control will be more of a problem than was originally thought, EPA contracted with IT Enviroscience, Inc., to investigate the various commercial sulfur-removal technologies and to propose a pilot-plant design based on the most cost-effective process for the removal of gaseous sulfur compounds from oil-shale retort gases.

#### B. OBJECTIVES

The objectives of this study were to determine the most applicable control technology for control of sulfur emissions from oil shale processing facilities and then to develop a design for a mobile slip-stream pilot plant that could be used to test and demonstrate that technology.

#### C. APPROACH

The work conducted included an in-depth evaluation of available gas characterization data from all major oil-shale development operations in the United States. Data gaps and inconsistencies were identified and corrected where possible through working with the developers and/or researchers in the field. From the gas characterization data, duty requirements were defined for the sulfur removal systems. It was found that oil-shale retorting processes fall into two broad categories: direct-fired-retort processes and indirect-heated-retort processes, each category having distinctly different duty requirements.

The overriding factor that separates the two categories of retorting processes and that is dominant in the application of desulfurization technology is the  $\rm CO_2/H_2S$  ratio of the gas produced from the retort. Those from direct-fired retorts have  $\rm CO_2/H_2S$  ratios that range from 76 to more than 165 and thus would require that the sulfur-removal process selectively remove  $\rm H_2S$  in the presence of large amounts of  $\rm CO_2$ . Indirect-heated retorts produce gases with  $\rm CO_2/H_2S$  ratios in the range of 4.3 to 5, which would allow a nonselective process to be used.

During this study it was determined by the EPA that the greatest immediate concern is control of sulfur emissions from direct-fired oilshale retorting processes and that the pilot-plant design should be applicable to these retorting methods. Since application of desulfurization technology to gases from direct-fired-retorting processes is more limiting, the screening of available process technologies was based on the duty requirements for those gases.

# D. RECOMMENDED AVAILABLE H<sub>2</sub>S CONTROL TECHNOLOGY The class of processes that remove H<sub>2</sub>S and CO<sub>2</sub> from fuel gases is generically called acid-gas removal or gas-sweeting processes. Removal of acid gases and/or other gaseous impurities from gas streams is accomplished either by direct chemical conversion of the acid gas to another compound that can be more easily separated from the gas, by absorption into liquid, or by adsorption on a solid. The large volumes of gas that must be processed in a typical oil-shale plant will limit the application of desulfurization technology to high-capacity, liquidphase processes. Since CO<sub>2</sub> is absorbed to some extent by all liquidphase processes, the high CO<sub>2</sub>/H<sub>2</sub>S ratio of the gas limits the selection to those processes that can selectively absorb sulfur compounds in the presence of large amounts of CO<sub>2</sub>.

Of those processes that selectively remove  $H_2S$  by direct conversion of the  $H_2S$  to elemental sulfur, the Stretford process is the most effective one. Of those indirect processes that selectively remove  $H_2S$  by separating the  $H_2S$  as a concentrated acid-gas stream, the following

processes were selected as the most effective in their separate process classifications: the Selectamine and the Adip processes, which use MDEA as the absorbent, the Benfield, the Selexol, and the Diamox processes. The Benfield and Selexol processes require the gas to be at high pressure and thus were eliminated since compression of the gas for the purpose of desulfurization could not be economically justified.

Except for the Diamox process, all the candidate processes are capable of removing  $H_2S$  down to about 10 ppmv. However, organic sulfur compounds, principally COS, which exist in only trace amounts in the gas, are not significantly removed or are only partly removed by the various processes. However, the presence of those compounds may reduce the overall effectiveness to 98%.

For desulfurization of gases from direct-fired oil-shale retorts the Stretford direct process is the most cost-effective system. For the model case used to evaluate the various processes the total estimated cost of sulfur removed by the Stretford process would be about \$0.50 per barrel of oil produced, which is less than half that projected for the best of the other processes evaluated.

The Claus process is used to recover sulfur from the acid gas produced by the indirect sulfur-removal processes. The large amount of CO<sub>2</sub> in the gas makes the best of the indirect processes only marginally capable of producing an acid gas rich enough in H<sub>2</sub>S for processing by the Claus process. Thus to apply these processes multiple stages of selective absorption would be required to handle the gas produced by many of the direct-fired retorts. The process, however, does not effectively remove COS. Only trace quantities of COS (less than 50 ppmv) have been found in gases produced by direct-fired retorts and thus an overall effectiveness of 98% or more is projected for the Stretford process. If the quantity of COS in the gas should be higher than indicated by the currently available gas characterication data the removal efficiency could be less than 98%.

The Stretford direct process, on the other hand, is only minimally affected by the quantity of  ${\rm CO}_2$  in the gas and therefore is adaptable to the full range of gases produced by direct-fired retorts.

#### E. DESIGN OF PILOT PLANT

The pilot-plant design is based on the current state-of-the-art technology for commercial application of the Stretford process. The maximum design capacity of the unit is  $28.3~\mathrm{sm^3m}$  of feed gas and  $6.6~\mathrm{Kg}$  of sulfur per hour. The plant should be capable of reducing the  $\mathrm{H_2S}$  content of the gas to 10 ppmv or less, and  $\mathrm{CO_2/H_2S}$  ratios as high as 200 to 1 should be possible.

The pilot plant is sized primarily to remove  $H_2S$  from oil-shale gas produced by direct-fired retorts. However, use of an ejector-venturi gas-scrubbing system affords wide gas turndown capability for the system. The pilot plant thereby is capable of operating on a slip stream from any of the currently proposed direct or indirect oil-shale retorting processes in the United States.

To properly function, the feed gas to the pilot plant must be 120°F or less, with most of the ammonia removed. A gas cooling column has been incorporated into the pilot design for cooling and removing the ammonia from the feed gas. The estimated cost of the pilot plant with all equipment, instruments, and controls, assembled on skid mountings as a complete and operable unit, is as follows:

Range	With Cooler	Without Cooler
High	\$520,000	\$338,000
Probable	400,000	260,000
Low	308,000	200,000

#### F. CONCLUSIONS

The Stretford direct gas desulfurization process may be the only currently available commercial process capable of effectively removing  $\rm H_2S$  from gases produced by direct-fired retorts. Application of the Stret-

ford process or of any other process to the treatment of these gases would extend the technology of the process into areas in which no analogous experience is available. Many questions need to be answered before the process can be applied with confidence to a full-scale commercial shale-oil production facility.

The principal areas of concern are as follows:

- 1. absorption of CO2 versus gas characteristics,
- capacity of the solution for absorbing sulfur versus gas characteristics,
- rate of by-product thiosulfate formation versus gas characteristics,
- 4. disposition of COS and other organic sulfur compounds in the feed gas,
- 5. effects of unsaturated hydrocarbons in the feed gas on process operation, life of the Stretford chemicals, and quality of the sulfur produced.

The viability of the oil-shale industry hinges on an environmentally compatible sulfur-removal process. Although there are no federal industry standards for emissions for the oil-shale industry at this time, the State of Colorado has enacted legislation that limits industry emissions to less than 0.3 lb of sulfur dioxide per barrel of oil produced and an equal amount per barrel of oil refined. To meet this standard at least 96% of the sulfur in the gas expressed as SO<sub>2</sub> would have to be removed.

The area of air pollution compliance that is of the greatest concern to industry and government is the Prevention of Significant Deterioration (PSD) requirements of the Federal Clean Air Act. This concept was enacted to prevent the addition of specified pollutants above a prescribed baseline value in specified air regions. Colorado adopted a more stringent plan, which limits the maximum level of sulfur dioxide in the air to an annual average of 10  $\mu g/m^3$ . Thus the maximum quantity

of shale oil that can be produced will be limited by the effectiveness of the sulfur emission control system used.

Unless the Stretford process can be demonstrated as an effective and reliable process for treatment of direct-fired oil-shale gases, industry may have to resort to combusting the gas first and then using less effective flue-gas disulfurization techniques. Because of the stringent PSD requirements any increase in sulfur emissions could result in reduction of the potential production capacity of the shale-oil industry.

#### II. INTRODUCTION

#### A. OIL-SHALE RESOURCE<sup>1</sup>

Recently, as the price of imported oil began to rise and gasoline exceeded \$1.00/gal, interest was renewed in obtaining oil from oil shale as part of the solution to the energy problem. The solutions to the environmental problems that were once a cost deterrent to commercialization have become more acceptable now that crude-oil prices are exceeding \$20.00 per barrel.

The consumption of oil in the United States in 1978 was about 6 billion barrels. It has been estimated that the known oil-shale resources that could be tapped by use of existing technology are equivalent to 600 billion barrels, and the total resources in the Green River Formation in the west have been estimated at 2 trillion barrels of oil, which would provide oil for 100 and 333 years, respectively, based on the 1978 rate of consumption. There are very sizable deposits of shale in the east and midwest, which are referred to as Devonian shale. These shales are leaner in oil than western (Eocene) shales; but, if they are incorporated into the resource estimate, the total amount climbs to about 28 trillion barrels. With a resource of this magnitude available, some people would ask why we import almost half of our oil supply. Part of the answer lies in the fact that the organic fraction of the shale is small and that, if the yield is assumed to be 24 gal/ ton of shale, about 1 cubic mile of rock would have to be processed to yield the oil we consumed in 1978. Another part of the answer is that sizable quantities of environmentally undesirable components could be released to the air and to surface waters.

#### B. ENVIRONMENTAL CONSTRAINTS<sup>1</sup>

Sulfur, one of the undesirable components, is contained in oil shale up to about 2%, and a typical shale in the Green River Formation may contain 0.7%. Partitioning studies<sup>2</sup>, indicate that somewhere between 16 and 30% is liberated to the gas that is generated, with the majority of the sulfur remaining with the spent shale. Under some circumstances

oil produced from shale might be considered to be a low-sulfur resource; in this case, however, where immense volumes of rock are processed, the potential for sulfur emissions is sizable. The uncontrolled sulfur emissions from an 8,000 m<sup>3</sup>/day oil shale retort installation could be 86--.59 tonnes per day. The resulting impact on the semiarid region in the west would be significant.

The National Ambient Air Quality Standard (NAAQS) for sulfur dioxide limits the annual average to 80  $\mu g/m^3$  of air. If this were the only regulation that industry had to comply with, all sulfur species would probably be oxidized to sulfur dioxide and some form of flue gas desulfurization would be used for control. However, the oil-shale industry must also be concerned with regulations dealing with the amount of a specific material that may be emitted by a specific source. Although there are no federal industry standards for emissions for the oil-shale industry at this time, the State of Colorado has enacted legislation that limits industry emissions to less than 0.3 lb of sulfur dioxide per barrel of oil produced and an equal amount per barrel for oil refined. The area of air pollution compliance that is of the greatest concern to industry and government is the Prevention of Significant Deterioration (PSD) requirements of the Federal Clean Air Act. This concept was enacted to prevent the addition of specified pollutants above a prescribed baseline value in specified air regions. Colorado adopted a similar plan, but made the acceptable levels for PSD more stringent. Table II-1 summarizes the PSD requirements for the federal government and the State of Colorado, which puts limits on sulfur dioxide to an annual average of 10 µg/m3. The future potential of the oil-shale industry therefore hinges on the effectiveness of the sulfurremoval process.

#### C. PURPOSE OF STUDY

Oil-shale developers are involved in a number of significant pilotscale activities for the development of process technology, and have indicated to the EPA their willingness to cooperate on joint projects for sulfur control technology evaluation. To capitalize on this oppor-

Table 1. Maximum Concentration of SO<sub>2</sub> to Meet Colorado and Federal PSD Requirements

	SO <sub>2</sub> Maximum Concentration	(µg/m <sup>3</sup> )
	Colorado	Federal
Class I		
Annual average	2	2
24-hr maximum .	5	8
3-hr maximum	25	.25
Class II <sup>b</sup>		
Annual average	10	20
24-hr maximum	50	,91
3-hr maximum	300	512

Class-I areas are listed as national parks and as natural wilderness and primitive areas.

 $<sup>^{\</sup>mathrm{b}}$ Class-II areas are the rest of the country.

tunity and to explore the possibility that sulfur emission control will be more of a problem than was originally thought, EPA contracted with IT Environcience, Inc., to propose a pilot-plant design based on the most cost-effective process for the removal of gaseous sulfur compounds, as well as one that is mobile. This type of design would necessitate investigation of various commercial sulfur-removal technologies and their applicability to oil-shale gases.

#### D. APPROACHES AND LIMITATIONS

1. Alternatives for Control of Sulfur Emissions

Two approaches can be taken to control sulfur emissions: to remove sulfur compounds from the gas before it is combusted (a gas-sweetening process), thereby producing a sulfur-free fuel; or to first combust the gas and then remove the resulting SO<sub>2</sub> from the flue gases [flue-gas-desulfurization (FGD) process].

The latter approach is generally less effective, because most of the flue-gas scrubbers can remove only 90 to 95% of the sulfur in the flue gas. The volume of gases handled by the FGD process is much larger than that handled by the first process because the products of combustion are more voluminous than the fuel gas burned. Also, the FGD system generally consumes more water and chemicals and is more costly to operate than the gas-sweetening processes.<sup>4</sup>

Processes that remove sulfur from the fuel gas before it is combusted are generally more effective, with most processes capable of removing 98% or more of the  $H_2S$  in the gas. With these processes, however, sulfur in other forms, such as COS,  $CS_2$ , and mercaptans, may be only partly removed or not removed at all.

To meet the Colorado air quality control regulations of 0.3 lb/bbl of  $SO_2$  equalivant emissions, at least 95% of the sulfur in the gas would

have to be removed. For these reasons this study was directed toward the more effective fuel-gas desulfurization techniques.

### 2. Basis of Process Selection

During this study it was determined by the EPA that the greatest immediate concern was control of sulfur emissions from direct-fired oilshale retorting processes and that the pilot-plant design should be applicable to these retorting methods. Therefore the screening of available process technologies was based on the duty requirements for desulfurization of direct-fired retort gases.

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<sup>\*</sup>When a reference number is used at the end of a paragraph or on a heading, it usually refers to the entire paragraph or material under the heading. When, however, an additional reference is required for only a certain portion of the paragraph of captioned material, the earlier reference number may not apply to that particular portion.

#### III. OIL-SHALE GAS CHARACTERIZATION

#### A. CHARACTERIZATION OF OIL SHALES

#### 1. Introduction

In characterizing the gas produced by the various oil-shale retort processes, it is necessary to have some understanding of the nature and composition of the organic and mineral materials that comprise oil shale and of how the materials become distributed among the resulting products. The oil in oil shale can be neither squeezed nor drained out at ambient or moderately elevated temperatures. When the oil shale is heated to temperatures in excess of 200°C and up to 500°C, the material is destructively distilled, producing recoverable oil and gas and leaving a spent mineral residue.

#### 2. Composition of Oil Shale

Oil shales having similar assays of oil, in gallons per ton (gpt), were found also to have a fairly narrow range of compositions, as can be seen in Table 2. 1,2 However, even in the Mahogany ledge, which is about 21 m thick, the assay of oil varies from 37.5 to 312.3 Lp tonne.

The mineral material, after being fired to an ash, has the composition shown in Table 3.  $^{1}$  The minerals in their natural form in oil shale are varied and complex, and on a point-to-point basis the pH can vary from 8.5 to 10.3

3. Composition and Nature of Organic Matter in Oil Shale

The organic matter in Green River oil shale has a significant amount of hydrogen (see Table 4). 1,4 The ratio of about 1.5 moles of hydrogen per mole of carbon enabled about two-thirds of the organic matter to be converted to oil during retorting. In contrast, high-volatility bituminous coal has a hydrogen-to-carbon mole ratio of about 0.9 to 1.3

The substantial amount of oxygen present in the organic material indicates the likely presence of carboxyl groups. It is also likely that

Table 2. Properties of Green River Mahogany Ledge Oil Shale

	Amount (wt % of	f raw shale)
Component	Stanfield <sup>a</sup>	Smith
Carbon, organic	12.4	12.21
Hydrogen	1.8	1.76
Nitrogen	0.4	0.37
Sulfur	0.6	0.70
CO <sub>2</sub> , mineral	18.9	17.92
Ash	65.7	-
Mineral content of spent shale	-	65.26
•	99.8	98.22

asee ref 1; analysis based on sample that yielded 115 L/tonne.

Table 3. Ash Composition of Oil Shale a

Component	Amount (wt % of raw shale)
sio <sub>2</sub>	27.8
Fe <sub>2</sub> 0 <sub>3</sub>	3.0
A1 <sub>2</sub> 0 <sub>3</sub>	8.6
c <sub>a</sub> o	15.1
M <sub>0</sub> 0	6.5
s0 <sub>3</sub>	1.2
Na <sub>2</sub> 0	2.0
к <sub>2</sub> 0	1.5

aOil assay, 115 l/tonne (see ref 1).

bsee ref 2; analysis based on sample that yielded 105 l/tonne.

Table 4. Composition of Organic Matter in Green River Oil Shale

	Amount (s	wt % of raw shale)
Component	Stanfield <sup>a</sup>	Smith <sup>b</sup>
Carbon	76.5	80.5
Hydrogen	10.5	10.3
Nitrogen	2.5	2.4
Sulfur	1.2	1.0
Oxygen	9.3	5.8

See ref 1; analysis based on sample that yielded 115 l/tonne bsee ref 2; analysis based on sample that yielded 105 l/tonne

these carboxyl groups are saponified by the mineral material, whose pH after pyrolysis is about 11.5 Therefore only through pyrolysis, or destructive distillation, wherein organogenic CO<sub>2</sub> is released, can oil be recovered from oil shale. As further evidence of the change in the nature of the organic material, as implied by the term "destructive distillation," the oil produced from oil shale has a pour point of about 29°C, although oil will not drain from oil shale at this temperature.

The presence of sulfur and nitrogen in oil shale and shale oil is discussed below.

4. Recovery of Products from Pyrolysis of Oil Shale by the Modified Fischer Method

A quantity of oil shale representing 94 Lpt assay was obtained from 10 locations in the Green River formation and was blended to provide a quantity of uniform composition. Eight replicate pyrolysis runs were made by the modified Fischer method to determine the reproducibility of this method of assaying the yield of oil and gas from oil shale.<sup>2</sup>
Table 5 gives the quantity and composition of the gas produced during pyrolysis; the data, in material balance form, that were obtained are given in Table 6.

The recovery of materials expressed as percent of element contained in raw shale is shown in Table 7. These data show that the recovery of nitrogen in the oil and gas streams combined is only 53%, of which 8.6% is in the gas stream. They also show that the combined recovery of sulfur is only 25%, of which 16.9% is in the gas stream, and that if the shale oil were treated to remove nitrogen and sulfur the NH<sub>3</sub> content of the gas would increase about sixfold and the H<sub>2</sub>S content would increase by 50%.

5. Sulfur Chemistry Relating to Pyrolysis Gas

The oil shale in the Green River formations contains sulfur in two
major forms: organic (with -C-S- groups present) and pyrite, FeS2.

Table 5. Gas Produced by Pyrolysis of Oil Shale by Modified Fischer Method

	Amount of Gas Produced		
Component	(vol %) (DG) a	(sm <sup>3</sup> /tonne of raw shale) <sup>b</sup>	
CH <sub>4</sub>	17.44	4.7	
C <sub>2</sub>	5.44	1.5	
c <sub>2</sub> =	2.10	0.6	
c <sub>3</sub>	2.43	0.6	
c <sub>2</sub> c <sub>2</sub> = c <sub>3</sub> c <sub>3</sub> =	2.39	0.6	
$c_4^{3}$	1.50	0.4	
	0.60	0.2	
C <sub>A</sub>	1.20	0.03	
C <sub>4</sub> = C <sub>4</sub> C <sub>5</sub> ≡ C <sub>5</sub> = C <sub>6</sub> =	1.56	0.3	
C <sub>s</sub> =	1.45	0.4	
c <sub>e</sub> =	0.11	0.4	
co2	24.00	6.5	
co	5.09	1.4	
н <sub>2</sub>	29.36	7.9	
н <sub>2</sub> s	3.20	0.9	
NH3	2.12	0.6	
Total	100.00	27.03	

a<sub>See</sub> ref 2.

b Calculated gas production.

Material Balance on Pyrolysis of Oil Shale by Modified Fischer Method<sup>a</sup> Table 6.

	4 20 20 6	Retorted	Product (kg	Retorted Product (kg/tonne of raw shale)	ale)
Component	Raw Shale (kg/tonne)	Spent Shale	0i1	Gas	Water
Carbon, organic	121.9	28.7	80.3	14.3 <sup>b</sup>	
Hydrogen, organic	17.7	2.0	11.0	3.1	1.5
Nitrogen	3.8	1.8	1.7	0.3	
Sulfur	, 8°9	5.1	0.5	1.2	
Minerals	8.648	825.4	1.5	9.1	12.5
Totals	1000.0	863.0	95.0	28.0	14.0
	100%	86.3%	9.5%	2.8%	1.48

asee ref 2.

Dincluding 1.6 kg of carbon from mineral carbon.

Including mineral  $\infty_2$  and  $\mathrm{H}_2\mathrm{O}_2$ .

 $^{d}$ Determined by oxygen balance.  $^{e}$ Including oxygen as CO and  $^{CO}_{2}$ .

Table 7. Distribution of Elements in Products of Pyrolysis of Oil Shale a

	Distribution (wt %)				
Component	Spent Shale	Oil	Gas	Water	
Carbon, organic	23.6	65.9	10.5		
Hydrogen, total	11.4	61.9	18.0	8.7	
Nitrogen, total	46.6	44.8	8.6		
Sulfur, total	75.1	8.0	16.9		

asee ref 2.

The -C-S- groups are pyrolyzed in the presence of hydrogen to yield  $H_2S$ . Examination of the gas phase that is formed in this process indicated that it went to completion; but since there are still sulfur-containing organic compounds in the shale oil, the existence of small quantities of low-boiling organic sulfur compounds in the gas would not be surprising.

Pyrite can undergo hydrogenolysis to yield H<sub>2</sub>S as shown below:

FeS<sub>2</sub> + H<sub>2</sub> 
$$\longrightarrow$$
 FeS + H<sub>2</sub>S (pyrite) (hydrogen) (iron sulfide) (hydrogen sulfide)

Consequently one-half of the sulfur is available for recovery as  $H_2S$  and can appear in the gas. Sulfate salts, which are found in spent shales, are apparently formed in the combustion zone following pyrolysis. Table 8 shows the estimated quantity of sulfur in these two forms and the sulfur available for recovery (3.89 kg/tonne of raw shale). The sulfur recovered as  $H_2S$  (1.15 kg/tonne of raw shale; see Table 6) is estimated to be 30% of the sulfur available for recovery.

#### B. PARAHO RETORT GAS

#### 1. Introduction

Development Engineering, Inc. (DEI), a subsidiary of Paraho Development Corp., have developed an above-ground retort to the point that they are ready to design a commercial-size module. This retort can be designed to operate in either the direct mode, wherein combustion air is brought into direct contact with the shale in the retort, or in the indirect mode, wherein only recycled reheated pyrolysis gas is brought into contact with the shale. In the latter case the recycle gas is reheated by some of the pyrolysis gas being burned in a separate unit. Recently, DEI concluded that the direct mode is more energy efficient, and they are now emphasizing its development in preference to the indirect mode.

Table 8. Estimated Distribution of Sulfur in Oil Shale

	Distribution (1	kg/tonne of raw shale)
Sulfur Form	Total Sulfur	Available as H <sub>2</sub> S
Organic	1.52 <sup>b</sup>	1.52
Pyrite	4.73 <sup>c</sup>	2.37
Unaccountable	0.75d	!
Total	7.00	3.89

aoil assay, 104 l/tonne.

b<sub>Based</sub> on 0.45 kg of \$/36.6 kg of C; see ref 4.

CBased on 1.41 kg of S as pyrite/ 45 kg of S as organic; see ref 4.

 $<sup>^{</sup>d}$ Calculation based on 7.0 kg of total S per tonne of raw shale.

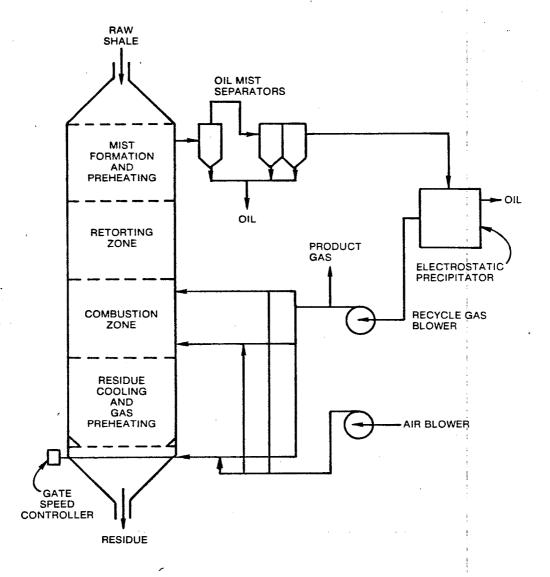


Figure 1. Flow Diagram for Paraho Direct-Mode Retort

DEI has submitted a proposal in response to a Department of Energy (DOE) Program Opportunity Notice (PON) for the design, construction, and operation of an above-ground retort process as part of DOE's plan to develop technology to exploit oil shale as an energy resource. The PON requires that such a retort process produce shale oil at the rate of  $960~\mathrm{m}^3\mathrm{PD}.^9$ 

# 2. Paraho Retort Process Description

In the direct mode of operation of the Paraho retort (see Fig. 1) crushed raw shale is fed to the top of the retort and spent shale leaves at the bottom. The shale passes downward by gravity successively through a mist formation and preheating zone, a retorting zone, a combustion zone, and finally, a residue-cooling, gas-preheating zone. The carbonaceous residue on the retorted shale is burned in the combustion zone to provide the principal fuel for the process. 10

The shale-oil vapors produced in the retorting zone are cooled to a stable mist by the incoming raw shale, and leave the retort at approximately 60°C. This mist is sent to a roughing cyclone, a condenser, and finally a wet electrostat precipitator, for oil separation. A portion of the cooled, oil-free, retort gas that remains is recycled to the retort. The remainder of the product gas is sent to the desulfurization train. 10

# Basis of Retort Design and Operation

It is understood that the Paraho retort will be designed according to the parameters shown in Table 9. The process will be operated on a sustained basis in order to prove the process and to develop engineering data on all streams. 9,11

# 4. Oil-Shale Composition

It is estimated that the oil shale to be used at the Paraho site will have the average composition discussed in Sect. III-A and summarized in Table 10. 4.9

Table 9. Design Basis for Paraho Retort a

Shale-oil production		
	960 m <sup>3</sup> /day	
Oil-shale throughput	10,000 tonnes/da	
Oil-shale assay	91.6—158 L/torn	
Method of heating .	Direct fired	
Gas recycle ratio	498 sm <sup>3</sup> /tonne	
Gas production ratio	223 sm <sup>3</sup> /tonne	
Gas mole ratio, CO <sub>2</sub> /H <sub>2</sub> S	76	
Heat content, HHV		
Pressure 3.4.3.	105,400 joules	
Temperature	Atmospheric	
a See refs 0 and 11	60°C	

See refs 9 and 11.

Table 10. Estimated Composition of Green River Oil-Shale for Paraho PON Retort<sup>a</sup>

Component	Amount (ct o)
Carbon, organic	Amount (wt %)
Carbon, mineral CO2	12.2
Hydrogen	4.9
Nitrogen	1.76
Sulfur, organic	0.37
Sulfur, pyrite	0.15
Sulfur, sulfate	0.47
Oxygen, except CO	0.08
Balance, minerals	1.37
Saturce, uniterats	_78.7
a	100.0

Oil assay, 104 to 112 l/tonne (see refs 4 and 9).

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As the pyrolysis gas is produced,  $498 \text{ sm}^3$  of the gas per tonne of shale is recycled with sufficient air to provide the necessary heat for pyrolysis, and  $223 \text{ sm}^3/\text{torme}$  is forwarded for uses external to the retort process. The gas produced is discharged at essentially atmospheric pressure and at about  $60^{\circ}\text{C}$  (see Table 11).

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Although the gas from the Paraho retort has been sampled, analyzed, and reported over the years, only the most recent data are considered as adequately representative of the process because the techniques and methods used now are considered to be better than those used before 1977. Table 11 shows the mean values of the composition of the gas produced by the Paraho retort in the direct mode during 1977 and 1978. In 1977, 31 gas samples were taken and analyzed and in 1978 there were 51. The high, low, and mean values for CO<sub>2</sub> H<sub>2</sub>S, and NH<sub>3</sub> are shown in Table 12. 12

- C. OCCIDENTAL VERTICAL, MODIFIED, IN-SITU-RETORT GAS
- 1. Introduction

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For more than six years Occidental Oil Shale, Inc. (OOSI) (formerly Garrett Research and Development Co.) has been engaged in modified, insitu, process technology development. Occidental's philosophy in developing this process for producing oil from shale is to maximize the recovery of in-place oil while minimizing costs and the environmental impact.<sup>13</sup>

OOSI prepared and processed its fifth and sixth retorts under a cooperative program with the Department of Energy (formerly the Energy Research and Development Administration). The DOE/Occidental Cooperative Agreement is a two-phase agreement consisting of engineering development of two specific retort designs for the Occidental modified in-situ process as the first phase and a technical feasibility demonstration as the second phase. 13

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Table 11. Gas Produced by Paraho Direct-Fired Retorta

		ount
Component	(sm <sup>3</sup> /tonne)	(dry vol %)
CH <sub>4</sub>	4.8	2.16
c <sub>2</sub>	1.2	0.56
c <sub>2</sub> .	1.5	0.66
c <sub>3</sub> 's	1.6	0.71
c <sub>4</sub> 's	-1.2	0.56
c <sub>s</sub> 's	1.2 <sup>b</sup>	0.57 <sup>b</sup>
co <sub>2</sub>	50.6	22.81
co	<sup>-</sup> 5.5	2.50
<sup>H</sup> 2	10.5	4.74
H <sub>2</sub> S	0,7	0.30
NH <sub>3</sub>	1.6	0.70
N <sub>2</sub>	141.5	63.80
02	0.2	0.09
50 <sub>2</sub>		(17 ppm)
NO <sub>X</sub>	•	(168 ppm)
CS <sub>2</sub> , RSH, COS		None detected
Thiophene <sup>a</sup>	•	(50 ppm)
Thio-4-(methylthio)- 3-cyclo-pentene-2-one <sup>c</sup>		(200 ppm)
Total, DG	222.2	100
H <sub>2</sub> O		20 <sup>d</sup>

aOil assay, 100 to 108 L/tonne; see ref 12.

 $<sup>^{\</sup>mathbf{b}}$  Total for  $^{\mathbf{C}}_{\mathbf{5}}$  and higher hydrocarbons.

c Identified sulfur compounds.

d<sub>20%</sub> H<sub>2</sub>O on wet-gas basis; see ref 9.

Table I2. Range of Critical Compositions in Gas from Paraho Retort by Direct Mode<sup>a</sup>

		Amount (dry vol %)	
Component	High	Low	Mean
co <sub>2</sub>	27.7	18.0	22.8
H <sub>2</sub> S	0.55	0.19	0.30
NH <sub>3</sub>	1.23	0.50	0.70

Represents 82 sample analyses taken in 1977 and 1978; see ref 8.

2. Description of Process

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The modified in-situ process consists of retorting a rubblized column of broken shale, formed by expansion of the oil shale into a previously mined-out void volume. In retort 5 the mined void volume is removed from the retort zone in the form of a vertical slice along the center of the room, extending from the top of the rubble pile to be formed, to the production level. This system is known as a vertical free-face retort system (VFFR). In retort 6 three horizontal levels are mined out similarly to room and pillar mining and then the oil shale is blasted above and below the mined-out sections to the horizontal rooms. This system is the horizontal free-face retort system (HFFR). 13

After the column of shale has been rubblized, connections are made to both the top and bottom and retorting is carried out (Fig. 2). Retorting is initiated by heating the top of the rubblized shale column with the flame formed from compressed air and an external heat source, such as propane or natural gas. After several hours the external heat source is removed and the compressed air flow is maintained, with the carbonaceous residue in the retorted shale used as fuel to sustain air In this vertical retorting process the hot gases from the combustion zone move downward to pyrolyze the organic matter in the shale below that zone, producing gases, water vapor, and shale-oil mist, which condense in the trenches at the bottom of the rubblized The crude shale oil and by-product water are collected in a sump and pumped to storage. Part of the off-gas is recirculated to control the oxygen level in the incoming air and the retorting temperature. The remainder of the off-gas passes on to the desulfurization train. 10

3. Basis of Retort Design and Operation
The projected bases for design and operation of Oxidental retorts 7 and
8 are given in Table 13. 14,15 The design basis for retort 6 is
also given.

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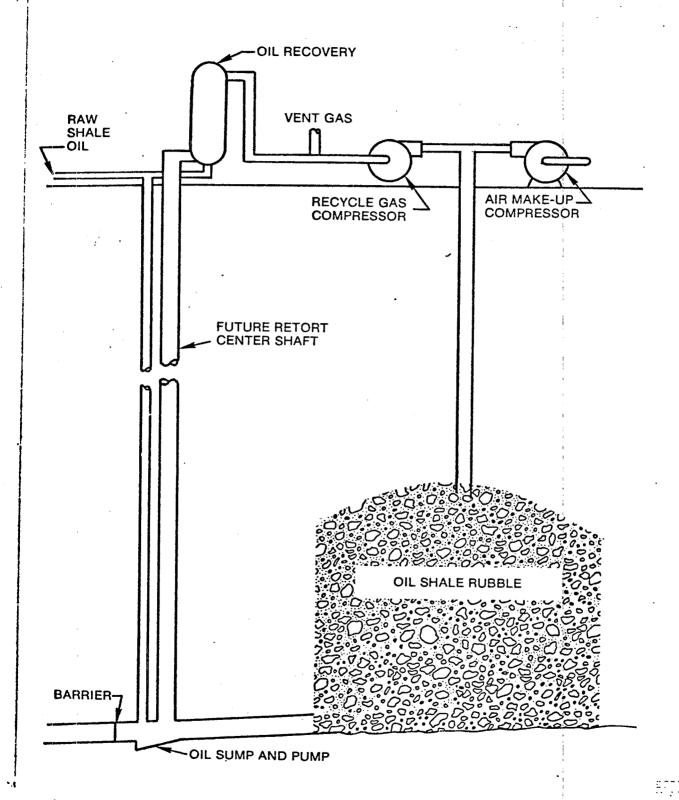


Figure 2. Flow Diagram of the Occidental Modified In-Situ Process

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Table 13. Design Basis for Occidental Vertical MIS Retorts

Parameter	Retort 6ª	Retorts 7 and 8
Retort size (tonnes)	329,058	618,120
Oil-shale assay (l/tonne)	62.5	62.5
Method of retorting	Direct in-situ	Direct in-situ
Gas injected	Air <sup>C</sup>	50% air/50% steam
Gas production rate (sm <sup>3</sup> /min)	408	1100
Gas productions ratio (sm <sup>3</sup> /tonne)	588	296
Gas ratio, CO <sub>2</sub> /H <sub>2</sub> S (mole/mole)	330 to 130	165
Heating value, HHV (joules/sm <sup>3</sup> )	$1.6 \times 10^6$ to $2.2 \times 10^6$	4.1x10 <sup>6</sup>
Pressure (psig)	Atmospheric	Atmospheric
Temperature (°C)	57 to 66	57 to 77

asee ref 13.

b<sub>Estimated</sub> from Oxy No. 6 and refs 14 and 15.

csteam injected at end of run.

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Retort-Gas Characterization

The composition of the gas produced by retort 6 at midpoint of the run is given in Table 14. The composition of the gas varied throughout In the initial stage of operation the CO<sub>2</sub>/H<sub>2</sub>S molar ratio the run. averaged about 330 to 1. After steady-state operation was established the CO<sub>2</sub>/H<sub>2</sub>S molar ratio averaged about 165 to 1.<sup>13</sup> Retorts 7 and 8 will be operated using a 50% oil/50% steam mixture injected into the retort. The use of steam will affect the composition of the gas produced. The quantity of COS in the gas is expected to increase.

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GEOKINETICS HORIZONTAL IN-SITU-RETORT GAS D.

Introduction

Geokinetiansis description dispersional of recovering shale oil from shallow formations of oil shale. The method consists of detonating implanted explosives to rubblize the shale. This raises the overburden, thereby developing void spaces among the broken shale. The bed of broken shale (retort) is then ignited with a fuel-air mixture injected at one end, and the retort products are expelled through a production hole at the other end (Fig. 3). The retort burn advances horizontally. When the retort is well ignited, the fuel is shut off and only air is fed to the injection hole.

Retort 18 was fired from November 16, 1979, to February 5, 1980. Retort firings will continue in an effort to develop this technology. The aim of this program is the production of 320 m<sup>3</sup> of oil per day over about a 10-year period from small shallow retorts.

Basis of Retort Design and Operation

 $^{\circ\circ}$  The design basis for geokinetics retort 18 is given in Table 15.16,17

3. Retort Gas Characterization

The average composition of the gases produced by retort 18 is given in

Table 16.

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Table 14. Gas Produced by Occidental Vertical MIS Retort

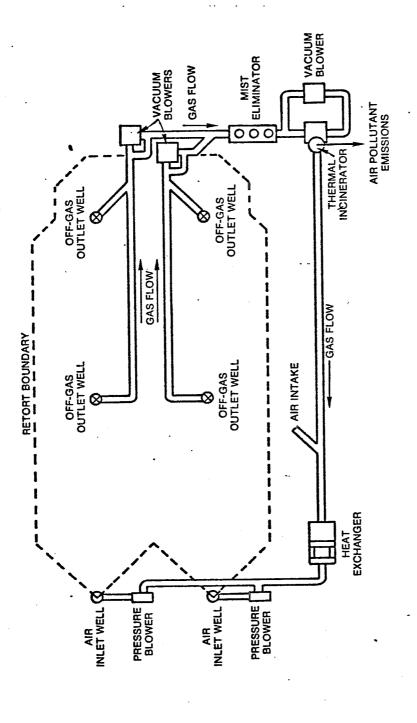
,	Compo	sition
Component	· (sm <sup>3</sup> /tonne <sup>b</sup> )	(vol %) (DG) C
H <sub>2</sub>	45.0	7.65
co	<b>5 . 2</b> ′	0.89
co <sub>2</sub>	188.8	32.26
CH <sub>4</sub>	8.4	1.44
C <sub>2</sub> H <sub>4</sub>	0.4	0.07
C <sub>2</sub> H <sub>6</sub>	1,3	0.224
C <sub>3</sub> H <sub>6</sub>	0.4	0.0
C3H8	0.6	. 0
c <sub>4</sub>	0.6	. 0
c <sub>5</sub> +	0.1	0.01
H <sub>2</sub> S	.0.7	0.1
NH3	 С	c
N <sub>2</sub>	330.5	56.41
o <sub>2</sub>	0.5	0.08
so <sub>2</sub>	Ò.9	0.15
NOX	0.2	0.03
Cs <sub>2</sub>	<b>c</b>	c
RSH	. <b>c</b> -	c
COS	0.02	(1-40 ppm)
Total, DG	585.9	99.611
H <sub>2</sub> 0		38.3 <sup>d</sup>

aGas produced by Oxy No. 6 at midpoint of run (Jan. 15, 1979); see ref 13.

b Average calculated values.

CNo data.

dwet-gas basis.



Flow Diagram of Geokinetics Retort No. 17 Off-Gas Handling System Figure 3.

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# الزير بر 15. Design Basis for Geokinetics Horizontal In-Situ Retort

Hat	10,025 b
Oll tonge)	83.3— 91.6
Oll *hale askay 3 * none)	Direct in-situ
Mailand of reterriti	Air
Unu injected  Unu production ====================================	45.3
Hay production - will	. c 181
Unu tatio, CO = (nole/mole)  Hamiling value, =//joules/sm3)	3,35x10 <sup>6</sup>
Planaure (psic)	Atmospheric
Tomperature (%)	54

Alim produce: 5 1/4/1/20ted at 20,134 m<sup>3</sup>.

National No. 15 52/1/4/20ted at 20,134 m<sup>3</sup>.

.6. Gas Produced by Geokinetics Horizontal In-Situ Retort<sup>a</sup> Table 16.

Component	Composition (dry vol %)
H <sub>2</sub>	7.47
co co	8.03
<b>2</b>	23.48
CH4	1.61
C <sub>2</sub> H <sub>4</sub>	0.085
C <sub>2</sub> H <sub>6</sub>	0.297
C <sub>3</sub> H <sub>6</sub>	0.09
C3H8	0.13
C <sub>4</sub>	0.15
C <sub>5+</sub> b	0.071
H <sub>2</sub> S	0.13
<sup>NH</sup> 3	0.060
N <sub>2</sub>	57.4
$\sigma_2$	1.13
so <sub>2</sub>	c
NO <sub>X</sub>	c
cs <sub>2</sub>	c
RSH .	c
cos	. (40 ppm)
Total	100.13
н <sub>2</sub> о	15.0 <sup>d</sup>

<sup>&</sup>lt;sup>a</sup>Gas produced by Geokinetics retort No. 18; see ref 16.

 $<sup>^{\</sup>mathbf{b}}\mathbf{c}_{6}$  to  $\mathbf{c}_{10}$  hydrocarbons present but not measured.  $\mathbf{c}_{\mathrm{No}}$  data.

dwet-gas basis.

#### UNION SGR-3 RETORT GAS Ε.

1. Introduction

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The Union Oil Company of California (Union) have been involved in oil shale activities for more than 50 years. The development of their retorting technology was initiated in the early 1940s, and several variations of a vertical-kiln retorting process, with upward flow of shale and countercurrent downward flow of gases and liquids, have been developed. Two variations are known as retort A and retort B. Union Oil now proposes to construct a 9090 tonnes/day demonstration plant, using the retort-B process, together with all the necessary auxiliary facilities. 10

2. Description of Process

> In the retort-B process (shown in Fig. 4 `) as the crushed oil shale flows upward through the retort it is met by a stream of hot (510 to recycle gas from the recycle gas heater flowing downward. The rising oil-shale bed is heated to retorting remperature by countercurrent contact with the hot recycle gas, resulting in the evolution of the shale-oil vapor and product gas. This mixture of shale-oil vapor and product gas is forced downward by the recycle gas and is cooled by contact with the cold incoming shale in the lower section of the retort The liquid level in the lower section is controlled by withdrawal of the oil product. The recycle and product gases are removed from the space above the liquid level. The product gas is first sent to a venturi scrubber for cooling and removal of heavy hydrocarbons by oil scrubbing. That portion of the product gas not recycled is then sent to the desulfurization train. Before the gas reaches the desulfurization train, it may be compressed and oil-scrubbed to recover additional hydrocarbons. 10

3. Basis of Retort Design and Operation

The basis for design for the Union SGR-3 retort is given in Table 17.18,19

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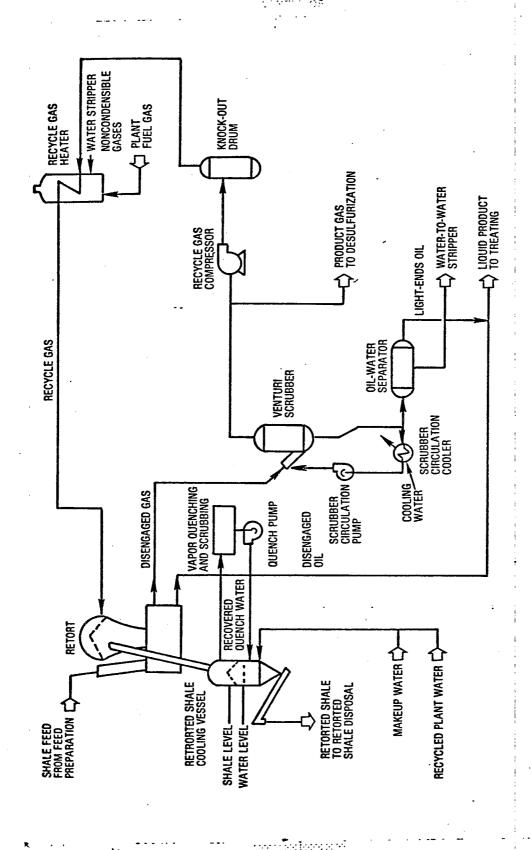
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Flow Diagram for Retort System in Union Oil Retort B Prototype Plant Figure 4.

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Table 17. Design Basis for Union SGR-3 Retorta

Shale-oil production		1440 m <sup>3</sup> /day <sup>b</sup>
Oil-shale throughput		9090 tonnes/day
Oil-shale assay		157.4 %/tonne
Method of retorting		Indirect mode
Method of heating		Indirect heating of recycled retort gas
Gas production ratio		29.9 sm <sup>3</sup> /tonne
Gas production rate		188.7 sm <sup>3</sup> /min
Gas mole ratio, CO <sub>2</sub> /H <sub>2</sub> S	<u> </u>	<b>4.3</b> 5
Heat content, HHV		$42.38 \times 10^6$ joules/sm <sup>3</sup>
Pressure		5-10 psig
Temperature (°C)		65.5—71.1°C

a See refs 18 and 19.

 $<sup>^{\</sup>rm b}$ Stripped shale-oil production is 1419  $^{\rm m}$ 3/day.

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4. Retort Gas Characterization

The estimated composition of the gas that will be produced by the Union SGR-3 retort is given in Table 18.

F. TOSCO-II RETORT GAS

1. Introduction

Tosco II is a process developed by the Oil Shale Corporation (Tosco). Initial development work began in 1955; in 1964 a 909 tonne/day semiworks plant was constructed and tested. A full-scale commercial plant is planned by the Colony Development Operation, a joint venture of several companies who formed or own Tosco.

2. Description of the Process

The heart of the processing sequence is the Tosco-II pyrolysis (retorting) unit and associated oil recovery equipment. The flowsheet for a single unit (or train) is shown in Fig. 5. The raw shale from the final crusher is fed to a fluidized bed, where it is preheated to about 500°F with flue gases from a ball heater. The residual hydrocarbons in the flue gases are burned in the ball heater. 10

The preheated shale is fed to a horizontal rotating retort (pyrolysis drum), together with approximately 1.5 times its weight in hot ceramic balls from a ball heater in order to raise the shale to pyrolysis temperature (482°C) and convert its contained organic matter to shale-oil vapor. The oil vapors are withdrawn and fed to a fractionator for hydrocarbon recovery. The mixture of balls and spent shale is discharged through a trommel, in order to separate the emerging warm balls from the processed shale.<sup>10</sup>

The warm balls are purged of dust with flue gases from a steam preheater, and the dust is removed from the flue gases by wet scrubbing.

The dust-free warm balls are returned to the ball heater via the ball elevator. In the ball heater they are reheated to about 704°C and then recirculated to the pyrolysis drum. The hot spent shale is cooled

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Table .18. Gas Produced by Union SGR-3
Indirect-Heated Retort

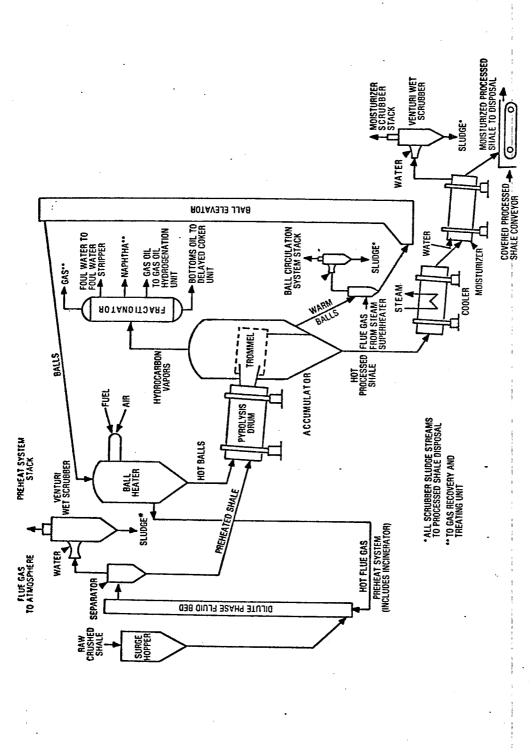
•	Amou	nt
Component	(sm <sup>3</sup> /tonne)	(dry vol %
H <sub>2</sub>	6.95	23.34
<del>.</del>	1.44	4.85
<sup>20</sup> 2	4.95	16.62
CH <sub>4</sub>	- 6.63	22.28
C2H2	0.06	0.19
C <sub>2</sub> H <sub>4</sub>	0.51	1.72
2 <sup>4</sup> 2 <sup>H</sup> 6	2.28	7.67
C3H6	1.13	3.81
3 6 C <sub>3</sub> H <sub>8</sub>	Ъ	ъ
C <sub>4</sub> H <sub>6</sub>	1.02	3.43
i-C <sub>4</sub> H <sub>8</sub>	0.09	0.29
n-C <sub>4</sub> H <sub>8</sub>	0.79	2.66
i-C <sub>4</sub> H <sub>10</sub>	0.34	1.14
n-C <sub>5</sub> H <sub>8</sub>	_ <b>0.</b> 09	0.29
i-C <sub>5</sub> H <sub>8</sub>	0.12	0.39
C <sub>5</sub> H <sub>10</sub>	0.06	0.19
C <sub>5</sub> H <sub>12</sub>	0,28	0.94
C <sup>6+</sup> .	1.84	6.19
н <sub>2</sub> s	1.14	3.82
NH <sub>3</sub>	· b	b
N <sub>2</sub>	0.03	0.11
$o_2^z$	c	c
so <sub>2</sub>		(125 ppm)
NO <sub>X</sub>	ъ	b
cs <sub>2</sub>	•	(20 ppm)
RSH	: <u>-</u>	(164 ppm)
COS	0.02	(542 ppm)
Total, DG	29.77	100.00
н <sub>2</sub> о	5.95	20 <sup>đ</sup>

Oil assay, 157g/tonne;

b<sub>No data.</sub>

<sup>&</sup>lt;sup>C</sup>No data; likely nil.

 $<sup>\</sup>mathbf{d}_{\mathtt{Estimated}}$  value, wet-gas basis.



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Flow Diagram for Pyrolysis and Oil Recovery Unit TOSCO-II Process Figure 5.

to about 149°C in a rotating drum cooler and moisturized with water recovered from the plant's foul-water stripper unit. 10

Unlike some U.S. oil-shale facilities, the Tosco-II/Colony commercial plant will be designed not only to produce shale oil but also to upgrade it on-site to produce synthetic crude oil and liquid petroleum gas (LPG), with ammonia, sulfur, and coke produced as by-products. The shale-oil hydrocarbon vapors from the pyrolysis drum are separated into water, gas, naphtha, gas oil, and bottom oil in a fractionator. The water is sent to a foul-water stripper, the gas and naphtha are sent to a gasoline recovery and treating unit, the gas oil is sent to a hydrogenation unit, and the bottoms oil is sent to the delayed coking unit. 10

3. Basis of Retort Design and Operation

The basis for design and operation of the Tosco-II retort is given in Table 19. 20,21

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4. Retort Gas Characterization

The average composition of the gas produced by the Tosco-II retort is given in Table 20.

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Table 19. Design Basis for Tosco-II Retort<sup>a</sup>

Shale-oil production	$8,872 \text{ m}^3/\text{day}^b$
Oil-shale throughput (6 retorts)	60,000 tonnes/day
Oil-shale assay	158 £/tonne
Method of retorting	Indirect mode
Method of heating	Pyrolysis by heated balls
Gas production ratio	38.5 sm <sup>3</sup> /tonne <sup>c</sup>
Gas production rate (single retort)	267.6 sm <sup>3°</sup>
Gas mole ratio, CO <sub>2</sub> /H <sub>2</sub> S	4.95
Heat content, HHV	$35.26$ x $10^6$ joules/sm $3^d$
Pressure	Slightly positive
Temperature	60°C.

a See refs 20 and 21.

 $<sup>^{\</sup>rm b}$  Based on pilot study; see ref 21. Production as low sulfur distillate is 7,360  $^{\rm m}$ /day; see ref 20.

<sup>&</sup>lt;sup>C</sup>Calculated from data in ref 21.

 $<sup>^{\</sup>rm d}$ Calculated valve based on  ${\rm C}_{\rm 5}$  and higher MW hydrocarbons removed.

Table 20. . Net Gas Produced by Tosco-II Retort a

	Amou	nt
Component	·(sm <sup>3</sup> /tonne)	(dry vol %)
H <sub>2</sub>	7.74	20.2
co	1.31	3.40
co <sub>2</sub>	: <b>7.8</b> 3	20.38
CH <sub>4</sub>	· 7.75	20.2
C <sub>2</sub> H <sub>2</sub>	b	<b>b</b>
C <sub>2</sub> H <sub>4</sub>	3.21	8.39
C <sub>2</sub> H <sub>6</sub>	2.91	7.61
C <sub>3</sub> H <sub>6</sub>	. 2.73	7.14
С <sup>3</sup> Н <sup>8</sup>	1.28	3.35
C <sub>4</sub>	2.00	5.22
C <sub>5</sub>	c	c
H <sub>2</sub> S	1.58	4.12
NH <sub>3</sub>	đ	đ
N <sub>2</sub>	ъ	ъ
0 <sub>2</sub>	ъ	b
so <sub>2</sub>	ъ	b
NO <sub>X</sub>	b	Þ
CS <sub>2</sub>	Nile	Nile
RSH		(35 ppm)
cos	= <u>-</u>	(135 ppm)
Total	38.38	100
H <sub>2</sub> O	9.59	20 <sup>b</sup>

a See ref 21.

b<sub>None</sub> reported.

CReported in oil recovered.

 $<sup>\</sup>mathbf{d}_{\mathsf{Absorbed}}$  in water phase and does not appear in gas.

<sup>&</sup>lt;sup>e</sup>See ref 22.

<sup>\*</sup>Wet-gas basis.

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#### TREATMENT TECHNIOUES A.

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The class of processes that remove sulfurous compounds and carbon dioxide from gases is generically called acid-gas removal or gas-sweeting processes. Removal of acid gases and/or other gaseous impurities from gas streams is accomplished by chemical conversion to another compound, by absorption into a liquid, or by adsorption on a solid.

In the first method the gas is passed through a fixed bed or is contacted with a liquid, whereby the impurities are converted to another compound that can be more easily removed. Conversion is either by direct chemical reaction with the bed or liquid or by catalytic reaction.

In the second method the gas stream is contacted with a liquid, and the gaseous impurities are either chemically or physically dissolved in the liquid absorbent. The absorbent is subsequently regenerated to strip the absorbed gas and is then recycled.

When the method of adsorption is used on a solid, the gas is passed through a fixed bed of granulated solid material. The impurities are adsorbed and held by the solid adsorbent. When the bed becomes saturated, it is replaced or regenerated.

Numerous processes for removal of sulfur compounds have been developed from these methods and more than 30 of them have achieved commerical importance. Each commercial process has specific advantages, disadvantages, or limitations and was developed to satisfy specific needs. Some processes are designed to remove both H<sub>2</sub>S and CO<sub>2</sub>, whereas other processes are more selective toward H<sub>2</sub>S or CO<sub>2</sub> removal. All liquidphase processes will remove CO<sub>2</sub> to some degree, along with the H<sub>2</sub>S. Some processes are economical for only bulk removal of acid gas and will not achieve a high degree of acid-gas removal. Other processes will achieve a high degree of acid-gas removal, to the ppm level, but

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are not economical for gases containing large amounts of acid gases. Still other processes are effective only if the  $\mathrm{CO}_2/\mathrm{H}_2\mathrm{S}$  ratio is small, whereas others are more effective if the  $\mathrm{CO}_2/\mathrm{H}_2\mathrm{S}$  ratio is high. Some processes cannot be used at a low pressure, but some are equally effective at all pressures. Most processes must be carried out at nearambient temperatures; others require cooling or chilling of the feed gas; some can be used at elevated temperatures, whereas others must be used at elevated temperatures.

All H<sub>2</sub>S removal processes can be divided into the general categories of direct- and indirect-conversion processes. In the direct-conversion processes the sulfur compounds are directly oxidized to elemental sulfur or another compound that can be separated and recovered. In the indirect conversion processes the acid-gas components are removed from the feed gas and recovered as a separate stream, which is subsequently processed for recovery of the sulfur. Sulfur is recovered from the concentrated acid-gas stream by the Claus sulfur recovery process or by one of the direct-conversion processes. Both these categories can be further broken down into either gas-phase dry-bed processes or liquid-phase (wet) processes (see Fig. 6).

- 1. Direct-Conversion Processes
- a. Dry Bed—Several dry-bed processes have been described for direct removal of H<sub>2</sub>S from hydrocarbon gases based on the classic Claus reaction of H<sub>2</sub>S and SO<sub>2</sub> to form elemental sulfur. The application of these processes is limited because of possible plugging of the beds by condensable hydrocarbons and/or other impurities in the gas. None of these processes are known to have achieved commercial importance.

The Claus process, while a dry-bed process, is not in the true sense a gas-treating or sulfur-removal process and therefore is not included in this discussion. The Claus process is used to recover sulfur from hydrocarbon-free acid-gas streams containing large amounts of H<sub>2</sub>S and is discussed in Sect. IV-B.

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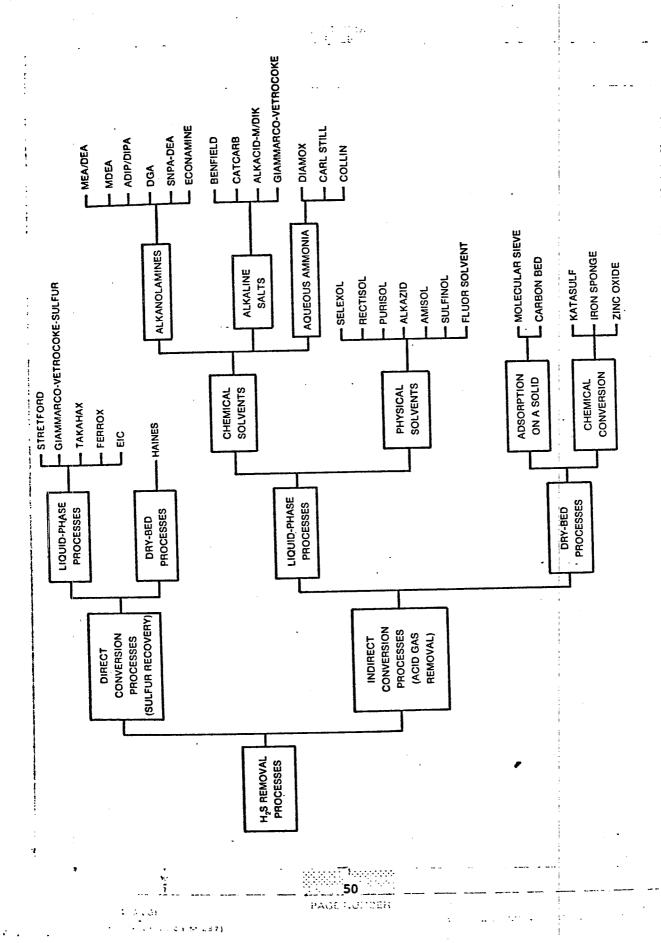


Figure 6. Gas-Sweetening Processes

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Liquid-Phase Direct-Conversion Processes—Liquid-phase direct-conversion processes usually are best suited to gases containing low concentrations of H<sub>2</sub>S, especially in the presence of substantial concentrations of CO<sub>2</sub>. Very little CO<sub>2</sub> is absorbed by the solution, and thus the processes selectively remove the H<sub>2</sub>S. The principal disadvantage is the relatively low capacity of the solutions for H<sub>2</sub>S, which results in large liquid circulation rates and requires large equipment for separating and processing precipitated sulfur. Generally plants of this type are not economical when the sulfur production rate exceeds 9 tonnes per day. Another disadvantage is that the considerable reaction heat generated by oxidation of H<sub>2</sub>S has to be dissipated at low temperature and cannot be recovered.<sup>2</sup> The quality of sulfur produced is lower than that of the sulfur obtained with the indirect processes, and the wastewater from purge streams is a major problem.

Direct-conversion liquid processes have substantial economic advantages over direct-conversion dry-bed processes in that they require less space, eliminate the high cost of bed replacement, and produce higher quality sulfur. Some liquid processes are capable of producing a treated gas of high purity equal to that obtained with dry-bed processes.

- 2. Indirect-Conversion Processes
- a. Dry Bed—The dry-bed indirect-conversion processes are generally applied on a batch-loaded basis, where the bed is removed from service when it is loaded and the sorbent is replaced or regenerated. The dry-bed processes are selective for H<sub>2</sub>S and generally achieve a high degree of removal. These processes are limited to the treatment of gas streams of small volume or to gas streams containing relatively low amounts of sulfur compounds. Regeneration of the bed is only partly achieved, and the bed eventually becomes plugged with sulfur and must be replaced. The economics generally limit application of these processes to plants recovering less than 10 tons of sulfur per day.<sup>3</sup>

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The advantages of dry-bed processes are simplicity, low cost, and ease of operation. Essentially complete removal of H<sub>2</sub> can be achieved, and some processes will selectively remove H<sub>2</sub>S without removing any CO<sub>2</sub>. The disadvantages are the large space requirements for the equipment, high maintenance costs for bed replacement, and problems with sulfur recovery.

b. <u>Liquid-Phase Process Involving Chemical Solvents</u>—Chemical solvents involve absorption by reversible chemical reaction of the H<sub>2</sub>S with a water solution of the absorbent. The following three types of chemical solvents are used:

Alkanolamines -- Many types of alkanolamines are used as absorbents (Fig. 6 ), and numerous processes and variations of them have been applied, depending on the physical conditions of the gas to be treated, its composition, and its purity requirements. Generally alkanolamine processes exhibit high reactivity with H2S and achieve a high degree of The processes are not sensitive to pressure, and operate at mear-ambient temperatures. Organic sulfur compounds are removed to some extent, but the lighter molecular weight amines may form nonregenerative compounds when absorbing organic sulfur or cyanide compounds. The lighter amines, though less expensive and capable of higher absorption capacity, are not selective in absorbing H2S, present corrosion problems in stripper and heat-exchanger surfaces, and are plagued by higher solvent vaporization losses than are the heavier amines. Alkanolamine processes are generally preferred for treatment of gases containing moderate amounts of acid gas at near-ambient temperature and pressure.

Alkaline salts—Alkaline salt processes employ an aqueous solution of a potassium or sodium salt that forms a buffered solution with a pH of about 9 to 11. The weak alkaline solution chemically absorbs the acid-gas components. These processes are of two types: those carried out at low (ambient) temperature and pressure and those carried out at elevated temperature and pressure. The low-pressure processes have

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> given way to more efficient processes and are not currently used. hot processes have an advantage when the feed gas is hot (up to 149°C) and under pressure (<100 psig). Stripping is accomplished partly by flashing, and thus the energy required for regeneration is lower than that for amine systems. The alkaline salt solution does not degrade significantly, and only minimum purge and makeup are required. Organic sulfur compounds are removed by the process, and it can be made partly selective toward H<sub>2</sub>S removal. The alkaline salt processes are usually the best ones to use when bulk removal of CO2 is required.

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Aqueous ammonia—Aqueous ammonia processes are generally applied when the gas to be treated contains ammonia. The ammonia contained in the gas is absorbed simultaneously with the H2S and thus serves as the active agent in the absorber solution. No chemical additives are required. Ammonia can be recovered in addition to the acid-gas components. Under certain operating conditions H2S can be selectively removed. Removal of up to about 97% of the H2S can be achieved; however, organic sulfur compounds are only partly absorbed.

Liquid-Phase Processes Involving Physical Solvents Physical-solvent processes are based on H2S, CO2, and minor gas impurities being more soluble in certain anhydrous organic solvents than are the fuel-gas components, i.e., hydrocarbons and hydrogen. Since their solubilities increase with pressure, the processes generally require high pressure The solvent is regenerated by pressure reduction, gas stripping, or heat applied to produce a concentrated stream of absorbed gas.

Most physical solvents have higher solubility for H2S than for CO2, and a high degree of selectivity can be obtained with some processes. solubility of hydrocarbons, however, increases with molecular weight. Consequently hydrocarbons above C4 are also largely removed with the acid gas.

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The advantages of physical solvents over chemical solvents are lower heat and power consumption, greater removal of organic sulfur components, higher selectivity for H<sub>2</sub>S, no purge or wastewater streams, and little corrosion. The disadvantages are that a high-feed gas pressure is required, heavier hydrocarbons are absorbed, the solvent loss is high, and the solvent cost is high.

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Physical-solvent processes are most economical when the feed gas is available at high pressure. These processes are usually used for bulk removal of acid-gas constituents when the acid-gas impurities make up an appreciable fraction of the total gas stream, making the cost of removing them by heat-regenerable chemical solvents less attractive. When high-purity treated gas is required, additional solvent regeneration steps beyond simple flashing are required.

## B. PROCESS FOR RECOVERING SULFUR

The Claus sulfur process is the principal process used for recovery of sulfur from acid gases produced by the indirect-conversion processes. The Claus process is discussed in some detail since it is important that the operation, capabilities, and limitations of the process be understood before fuel gas desulfurization processes are described.

A flow diagram of a conventional Claus system is shown in Fig. 7. The basic Claus reaction is

$$2H_2S + SO_2 \xrightarrow{} 3S + 2H_2O$$

The conventional process is carried out in stages: the first stage is at high temperature in a thermal oxidation furnace, where one-third of the H<sub>2</sub>S is combusted to SO<sub>2</sub> in the presence of air; the reaction in the succeeding stages occurs at lower temperatures in the vapor phase by catalytic oxidation. Usually two or more catalytic stages are employed. Sulfur is recovered after each stage by the gas streams being cooled and condensed. The gas is then reheated to the operating temperature of the succeeding stage. When the acid-gas feed is mostly

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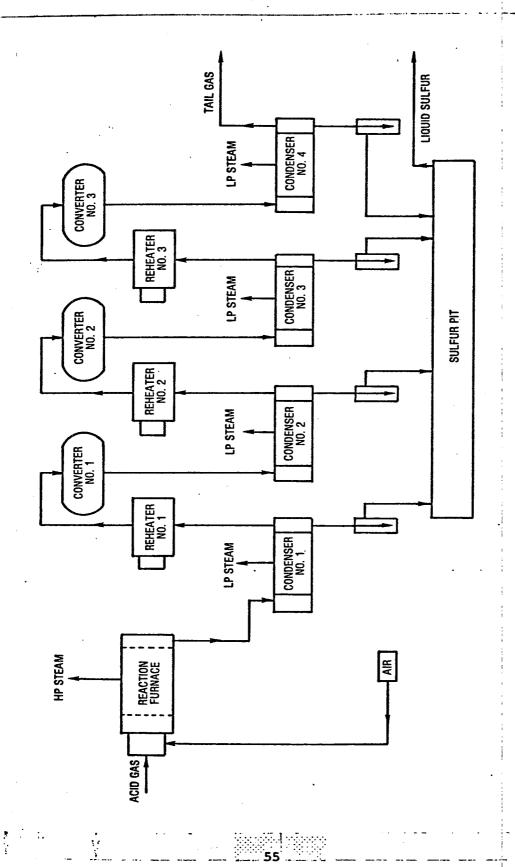


Figure 7. Typical Three-Stage Claus Process

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H<sub>2</sub>S, sulfur conversions of up to 70%<sup>4</sup> can be obtained by the thermal oxidation step alone, and overall conversions of up to 99%<sup>5</sup> have been obtained by use of four catalytic stages. The average sulfur recovery yield obtained with properly designed and operated systems usually ranges from 93—97%.<sup>6</sup> The remaining 3 to 7% of the sulfur is present in the Claus unit off-gas.

Recent developments have improved the process so that sulfur recovery levels of above 99% are routinely achieved by employing tail-gas treatment; 5 acid-gas streams containing as little as 5% H<sub>2</sub>S have been processed with modified Claus systems. 7

A necessary condition for successful operation of the thermal oxidation furnace is that the acid-gas mixture be rich enough (high in  $H_2S$  content) to ensure stable combustion at the required reaction temperature. The gas stream entering the first-stage catalytic converter should be at its stoichiomatic ratio of 2 parts of  $H_2S$  to 1 part of  $SO_2$  and be free of oxygenated impurities ( $SO_3$ ), unburned heavy hydrocarbons, soot, and excessive ammonia. An imbalance of either  $H_2S$  or  $SO_2$  will cause the excess component to pass through the system unreacted. Oxygen causes sulfonation of the catalyst, which results in loss of catalyst activity. Soot and carbonaceous material and ammonia, through formation of ammonium sulfate, cause the catalyst beds to become plugged. Unburned hydrocarbons that condense and stick to the catalyst will also cause loss of catalyst activity.

The operating temperature required for the thermal oxidation furnace is largely governed by the impurities in the acid gas. If the acid gas consists of  $H_2S$ , COS,  $CS_2$ , and  $CO_2$ , a furnace temperature of 700 to 800°C is sufficient. If the gas contains hydrocarbon compounds, ammonia, and other combustible impurities, temperatures of 1000 to 1200°C are required to assure complete combustion of the impurities and elimination of soot formation. The presence of combustible impurities in the acid gas increases the amount of air that must be introduced to

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the combustion chamber, and their products of combustion greatly add to the quantity of process gas entering the catalytic stages.

Methods for Processing Acid Gases Containing 16 to 100% H2S If the acid gas contains 50 to 100% H2S, the total volume of acid gas is normally fed to the combustion chamber (the classic, or the straight-through, process). The amount of air fed to the reaction furnace is controlled to combust one third of the H2S and all the ammonia and hydrocarbon impurities in the feed. If the temperature is maintained at the proper level (1000 to 1100°C), organic impurities are oxidized and the catalytic stages operate without significant problems. As the HoS content of the acid gas decreases, the lower the temperature obtained in the reaction furnace becomes. Longer residence time in the furnace is required to compensate for this lowered temperature, and thus larger chamber volumes are needed to maintain the dyanmic equilibrium. Below 35% H2S, the combustion required to produce the SO2 for the Claus reaction is no longer possible without special modifications being employed.

Below 50% H2S, the Claus process is usually modified. If the H2S content is between 16 and 50%, the split-flow process is often applied. With this process only one-third of the acid gas is fed to the thermal oxidation furnace. Stable combustion is obtained by the H2S being completely combusted to SO2. The other two-thirds of the acid gas bypasses the furnace and is mixed with the gases from the thermal oxidizer before they enter the first catalytic converter. However, impurities in the acid-gas stream are detrimental to the downstream catalytic section of the Claus plant and can affect the service life of the catalyst and the purity of the sulfur produced. If impurities are present in the acid gas, the complete stream should be fed to the combustion chamber as in the straight-through process. Preheating the combustion air and/or acid-gas stream or burning supplemental fuel or sulfur will enable stable combustion to be obtained.

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- 2. Methods for Processing Acid Gases Containing Less Than 16% H<sub>2</sub>S

  If the acid gases contain less than 16% H<sub>2</sub>S (called lean acid gases),
  the methods for obtaining stable combustion involve the following processes:
- before it enters the thermal oxidizer. Preheating is accomplished by heat exchange with the hot gases exiting the combustion chamber, by steam or by externally fired heaters. Only a limited amount of heat can be obtained by preheating. For lean acid gases it may be nessary to use supplemental fuel in direct-fired burners. However, burning the fuel adds to the cost of operation and to the volume of gases that must be handled by the downstream components.
- b. Sulfur Burning—By-Pass Process—The Claus thermal oxidizer as such is eliminated. The SO<sub>2</sub> required for the reaction is obtained by burning liquid sulfur in a separate combustion chamber. This has the advantage that stable combustion is obtained in the sulfur burner and that the gas produced is mostly SO<sub>2</sub>, depending on the purity of sulfur burned. Off-grade sulfur can be burned since the carbonaceous material is oxidized to carbon dioxide and water. The acid-gas stream is preheated in a gas-fired heater. The hot acid gas and the SO<sub>2</sub> from the sulfur burner are then mixed and fed to the catalytic converter. Application of this process is limited to relatively pure acid gases since even small quantities of hydrocarbon can cause carbon and tar to deposit on the catalyst and/or the sulfur quality to be degraded.
- c. Oxygen Process—Oxygen in place of air is used in the thermal oxidizer. This eliminates the ballasting effect of the nitrogen; thus less heat is required to maintain equilibrium and the volume of gases passing to the catalytic stages is reduced. Depending on the quantity of combustible material in the feed, oxygen is mixed to the acid gas or to a portion of the acid gas, which is then fed to one or more burners. If insufficient heat is generated because of lean-acid-gas mixtures, the supplemental fuel must be burned.

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#### 3. Cost Data

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Table IV-1 gives the relative capital and operating costs of the various process options, based on cost data reported by Fischer.8 The sulfur-burning by-pass process results in the lowest capital and operating costs of all processes available for processing lean acid gases. However, its application is limited to relatively hydrocarbon-free acid gases. When the H2S and hydrocarbon contents are low, the oxygen process is preferable provided that the oxygen can be obtained at low cost. The preheating process is used for those applications in which the acid gas contains hydrocarbon impurities and the  ${\rm H}_2{\rm S}$  content is moderately low.

The sulfur-burning straight-through process is the most expensive proc-Since the entire gas stream is introduced to the combustion chamber, its temperature must be raised to the furnace operating temperature. The higher the furnace operating temperature, the higher the operating cost and the larger the equipment required.

4. Claus Tail-Gas Treatment Processes

> The tail gas from a properly designed and operated Claus sulfur recovery plant may contain from 8,000 to 28,000 ppmv of sulfurous compounds (H<sub>2</sub>S, SO<sub>2</sub>, COS, CS<sub>2</sub>, S vapor). In order to comply with the more stringent current regulations, many processes have been developed to clean up Claus tail gases.

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Table 21. Comparison of Claus Process Options<sup>a</sup>

	Rela	Relative Gas Volume	d amu	Rele	Relative Capital Cost	Cost	Relat	Relative Operating Cost	g Cost
Option	At 900°C	At 900°C At 1000°C At 1100°C At 900°C At 1100°C At 900°C At 1000°C At 1100°C	At 1100°C	At 900°C	At 1000°C	At 1100°C	At 900°C	At 1000°C	At 1100°C
process	1.00			1.00			1.00		
Oxygen Drocess	0.98	90.1	1.13	1.66	1.97	2.50	1,26	1.50	1.92
Probeat process	1.32	1.53	1.91	1.77	2.06	2.56	1.80	2.11	29.62
Sulfur burning—straight-through process	1.69	1.96	2.45	2.16	2.59	3.13	1.95	2.34	2.82

ansed on cost data reported in ref 8.

 $<sup>^{</sup>m b}_{
m Gas}$  volume to first catalytic reactor for feed gas containing 8%  $_{
m H_2}{
m S}.$ 

Thermal oxidizion stage eliminated, gas is indirectly heated only to operating temperature of first catalytic stage.

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shows the principal Claus tail-gas treatment processes. Three basic approaches to tail-gas cleanup are employed. With the first approach all sulfurous compounds are reduced to H2S in a hightemperature catalytic converter. The gas is then cooled and the condensed water removed. After the H2S is cooled, it is either absorbed and subsequently regenerated as a concentrated acid-gas stream for recycle back to the Claus unit or is reacted and recovered as elemental sulfur. With the second approach all sulfurous compounds are oxidized to  $SO_2$  in a thermal oxidation furnace. The  $SO_2$  is then either absorbed and subsequently regenerated as a concentrated SO<sub>2</sub> gas stream for recycle back to the Claus unit or is reacted to form another compound, which is then recovered as by-product. The third approach is to extend the Claus reaction at lower temperatures. When a catalytic system is operated at a lower temperature than normal (near or below the sulfur condensation temperature), the thermodynamic equilibrium for the formation of sulfur from H2S and SO2 is further enhanced. Processes based on this approach are carried out in either the liquid phase with a catalyst present or a gas phase, in which a solid catalyst bed is used.

The choice of which tail-gas treatment process to use depends on many factors, principally those given below:

- 1. the composition and volume of the tail gas,
- whether the system can be integrated as part of a new installation or whether it is an add-on to an existing Calus plant,
- the initial investment cost and/or the operating costs,
- 4. the utilities required, the sulfur product produced, and/or the wastes produced,
- 5. the performance, reliability, and operating range of the system.

The Shell Claus off-gas treating (SCOT) process, the Beavon sulfury recovery process (BSRP), and the Wellman-Lord (W-L) process are the most widely applied processes. If the tail gas is low in CO<sub>2</sub>, the SCOT process usually produces the best economics of the three. If the gas

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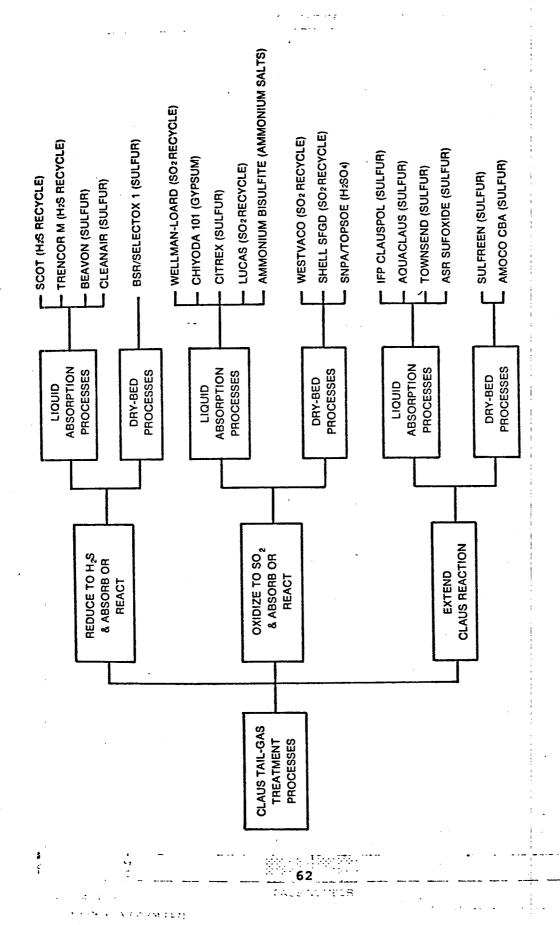
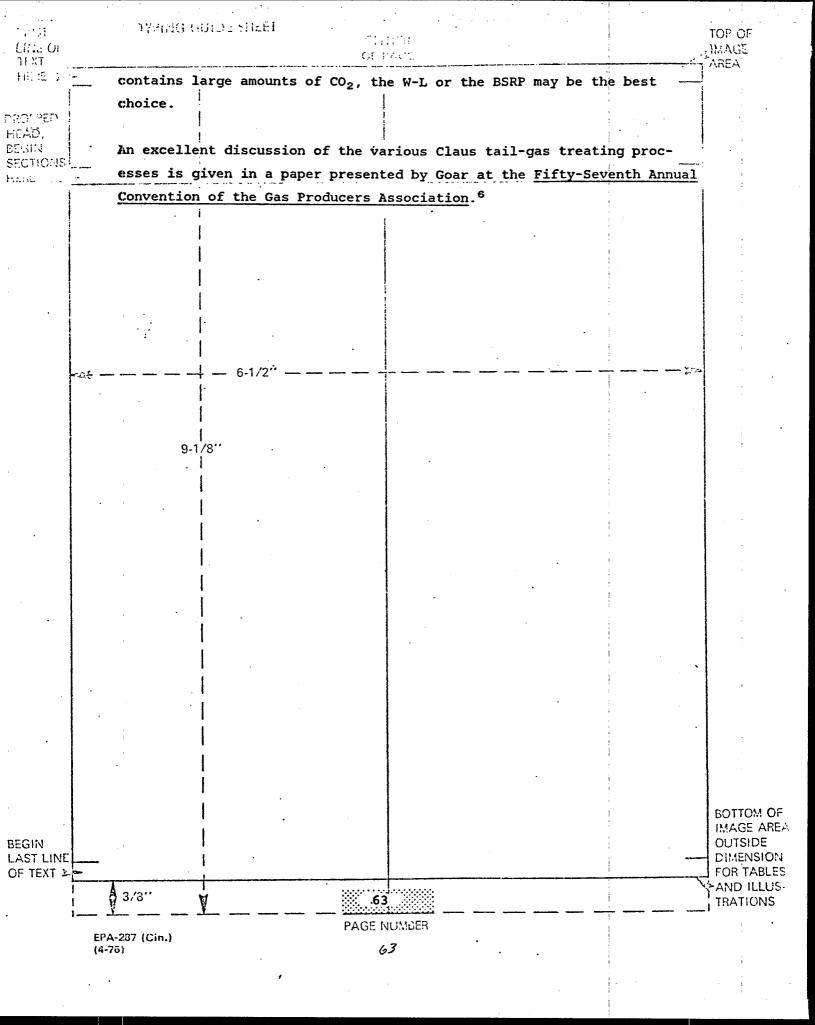


Figure 8. Claus Tail-Gas Treatment Processes



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#### FACTORS INFLUENCING CHOICE OF PROCESS

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In selecting a gas-treating process to remove sulfur compounds many factors have to be considered. The most important are the following:

- 1. the product-gas specifications.
- 2. the quantity of acid-gas components (CO2 and H2S) contained in the
- 3. the influence of impurities such as COS, CS2, mercaptans (RSH), NH3, and HCN,
- the quantity of heavier hydrocarbons in the gas, 4.
- the condition of the feed gas (temperature and pressure),
- the capital and operating costs of the process.

#### PRODUCT-GAS SPECIFICATIONS

#### Sulfur Compounds 1.

Sulfur compounds are removed to prevent air pollution by SO2 when the gas is combusted. They are also removed for safety and to prevent corrosion and odor problems. For natural-gas distributed by pipeline the total sulfur content is reduced 4.6 grams, or less, per 100 m<sup>3</sup> of gas. If the gas is used as chemical feed stock, sulfur compounds often have to be reduced to less than 1 ppm to protect sensitive catalyst systems.

Generally the cost of removing the sulfur increases as the degree of removal required increases. Achieving a high degree of removal often requires sacrificing other desirable process features, such as H2S selectivity, process simplicity, or favorable operating economics. higher the degree of sulfur removal required the more limiting the process selection becomes.

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·2. CO2

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Carbon dioxide and inert gases have no harmful effect other than that of reducing the heating value of the gas. Often CO2 can be tolerated and does not have to be removed.

3. COS, CS2, and RSH

> Carbonyl sulfide, carbon disulfide, and mercaptans contribute to the total sulfur content of the gas and generally must be removed to the same degree as H2S.

4. Dew Point

> In a saturated gas a drop in temperature or an increase in pressure will cause water to be condensed in the lines. Therefore the gas \_\_\_\_\_ should be precooled to the lowest temperature to be encountered in the system. Similarly, condensable hydrocarbons could condense out if the gas conditions dropped below their condensation point.

5. NH3

> If not removed, ammonia could be a problem in that NO compounds could be formed when the gas is combusted. Also, ammonia could be detrimental to some desulfurization processes.

ACID-GAS COMPONENTS

The most important factor in choosing a gas-treating process is the quantity of acid-gas components, CO2 and H2S, contained in the gas and the volume ratio between these two components. If the total quantity of sulfur contained in the gas is small, less than 9 tonnes/day, a drybed or direct-oxidation process may be more economical. If the quantity of sulfur is greater than 9 tonnes/day, an indirect process may be more economical.

Although the Claus process is preferable for recovering sulfur from concentrated acid-gas streams produced by indirect processes, it requires an acid-gas feed containing greater than 15% H2S for effective.

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operation. The higher the concentration of  $H_2S$  in the acid gas the more effective the Claus system will be.

Since all liquid-phase indirect processes absorb some  $CO_2$  along with the  $H_2S$ , an indirect process selective toward  $H_2S$  should be used if the  $CO_2/H_2S$  ratio in the feed gas is greater than 1 and is essential if the ratio is greater than 6. If it is necessary to remove the  $CO_2$  along with the  $H_2S$ , when the  $CO_2/H_2S$  ratio is high, a process selective toward  $H_2S$  removal should be used followed by a process that can remove bulk quantities of  $CO_2$ . If the acid-gas stream contains less than about  $S_0$   $S_2$  it is usually more economical to process the acid-gas stream by a liquid-phase direct-oxidation process than by the Claus process.

### C. INFLUENCE OF IMPURITIES

Most direct-recovery processes will not remove COS or CS<sub>2</sub> and will only partly remove mercaptans. The alkanolamine processes will remove organic sulfur compounds, but the lighter alkanolamines (MEA) and to some extent DEA form nonregenerable thiosulfates, which degrade the solvent when COS or CS<sub>2</sub> is present in the feed gas. Most alkaline salt processes remove COS and CS<sub>2</sub> by hydrolyzing them to H<sub>2</sub>S, but mercaptans are only slightly absorbed. The physical-solvent processes effectively absorb organic sulfur compounds.

If ammonia is not removed before the acid-gas absorption step, it will be absorbed with most processes and end up in the acid-gas stream.

Most ammonia in the gas can be removed with the condensates in the precooling step. Small amounts of ammonia can be tolerated by the Claus unit if the quantity of ammonia is small compared to the quantity of H<sub>2</sub>S in the acid-gas stream.

Hydrogen cyanide is largely co-absorbed with the other acid gases and, if the absorbent solution does not contain oxidants, will end up with the acid-gas stream. In the direct-recovery processes HCN is converted

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to stable thiocyanates, which will build up in the solution and require that a portion of the solution be purged.

If condensible hydrocarbons are absorbed, poor-quality black sulfur will be produced by direct-recovery processes. Hydrocarbons in the acid-gas feed to the Claus unit will increase the combustion air required and necessitate that the Claus thermal oxidizer be operated at a higher temperature to assure complete combustion of hydrocarbons. If the hydrocarbons are only partly oxidized, the catalyst beds can become plugged or a poor-quality sulfur can result.

Generally the alkaline salt and alkanolamine processes will not appreciably absorb hydrocarbons. The physical-solvent processes, however, will readily absorb hydrocarbons heavier than butane.

## D. CONDITION OF FEED GAS

Most processes are carried out at near-ambient temperatures (16 to 54°C). The physical-solvent processes are most effective at lower temperatures, and some require that the gas be chilled. In most of the currently applied alkaline salt processes elevated temperatures of up to 149°C can be used.

The physical-solvent processes and most alkaline salt processes require a high pressure, above 150 psig, to function efficiently. The alkanol-amine and direct-recovery processes are generally independent of pressure and can be used equally well over a wide range of pressures.

#### E. CAPITAL AND OPERATING COSTS

The bottom line in choosing a process is the capital and operating costs. Generally the more sulfur contained in the gas the less the cost on a unit basis (\$/tonne) to remove it. If sufficient sulfur compounds are present in the gas, an indirect-recovery process followed by a Claus sulfur recovery unit will provide the best economics. The physical-solvent processes are more economical when the acid-gas content is high or when the gas exists at high pressure. The alkaline

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salt processes have an advantage over other processes when the gas is at high pressure and high temperature and contains moderate amounts of acid gas. The alkanolamine processes are most economical when the gas is at low pressure and contains little  ${\rm CO_2}$ . For gases that contain little H<sub>2</sub>S or that have very high CO<sub>2</sub>/H<sub>2</sub>S ratios the direct-conversion\_ processes may be more economical than the indirect-conversion processes. If complete removal of  ${\rm H}_2{\rm S}$  is required and the gas contains little  $H_2S$ , the solid-bed processes may be the most economical ones.

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# VI. DUTY REQUIREMENTS FOR OIL-SHALE RETORT-GAS-DESULFURIZATION SYSTEMS

#### A. CLASSIFICATION OF OIL-SHALE RETORTING PROCESSES

From the standpoint of removing sulfur compounds from oil-shale retort gases all retort processes can be divided into two broad and widely separated categories: those in which direct-fired retorts are used either above or below ground (in-situ), where combustion occurs within the retort; and those in which indirect-heated retorts are used and the shale is retorted in the absence of air.

The application of desulfurization technology for gases produced from direct-fired oil-shale retorting processes is distinctly different from that required for natural gas, coke-oven gas, refinery gas, or gas produced by coal gasification or, for that matter, for gas produced from indirect-heated oil-shale retorting process. Typical compositions of the gases are shown in Table 22. 1—3 The gas from direct-fired retorts contains large amounts of inert components, over 63% N<sub>2</sub> and 22% CO<sub>2</sub>, but the H<sub>2</sub>S content is less than 0.3%; it also contains large amounts of ammonia and unsaturated hydrocarbons, such as acetylene, ethylene, propylene, butylene, and butadiene. More than 40 hydrocarbon compounds have been identified in the gas; it is saturated with water and contains some oxygen and trace amounts of sulfur species other than H<sub>2</sub>S.

Table 23 gives the range of gas compositions from direct-fired retorts, and Table 24 gives the range of those produced by indirect-heated retorts. The overiding factor that separates the two groups of processes and that will be dominant in selection of sulfur removal processes is the  $\rm CO_2/H_2S$  ratio. For direct-fired retorts the  $\rm CO_2/H_2S$  ratio ranges from 76:1 to more than 165:1, thus requiring a sulfur removal process that will selectively remove  $\rm H_2S$ . Indirect-heated retorts produce a  $\rm CO_2/H_2S$  ratio in the range of 4.3:1 to 5:1 which would allow a nonselective process to be used. As defined in Table 25

 ${
m H_2S}$  selectivity is the molar ratio of  ${
m H_2S}$  absorbed to  ${
m CO_2}$  absorbed. If an indirect recovery process were to be used in conjunction with a

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Table 22.. Comparison of Fuel Gases

Parameter         Direct-Pired Retort         Indirect-Route         Inugic         Kopperso-Goven         Gasd GasC         Refinery Gasd GasC         Natural GasC           Components, vol 4         0.3         4.1         1.1         1.6         0.3         62.5         2.6           My2         22.8         20.4         31.2         7.0         2.2         4.9         1.4           NH3         0.7         8.3         0.2         9.4         1.0         8.3         1.4           C2         1.2         16.0         0.7         0.7         8.4         82.3         1.4           C3         0.7         10.5         7         6.7         1.0         1.4         1.4           C4         0.6         0.7         0.7         0.7         4.6         1.8         1.8           C5+         0.6         5.2         39.1         32.7         52.4         4.4         0.8           M2         0.6         0.1         0.6         1.2         4.1         4.8           L2,4         0.1         0.2         3.4         17.3         57.4         4.8           Lamin         0.2         0.1         0.6         1.2		Oil-Shale Gas	le Gas	Coal Ga	Coal Gasification	Coke	ı	
omponents, vol % $ \begin{array}{ccccccccccccccccccccccccccccccccccc$	Parameter	Direct- Fired Retort	Indirect- Heated Retort	Lurgi	Koppers- Totzek	Oven Gasd	Refinery <sub>Gas</sub> c	Natural Gase
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	Components, vol %							
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	S.H	0.3	4.1	1.1	1.6	0.3	62.5	2.6
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	~ §	22.8	20.4	31.2	7.0	2.2	4.9	1,4
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	, NH,	0.7	8,3	0.2	· <b>-</b>	1.0		
$C_2$ 1.2       16.0       0.7       16.0       0.7       4.0       4.6 $C_4$ 0.6       5.2       0.7       4.6       10.0 $C_4$ 0.6       0.7       4.4       4.4 $C_5$ 4.7       20.2       39.1       32.7       52.4 $N_2$ 63.8       0.6       1.2       4.1 $N_2$ 0.1       0.6       0.1       0.1 $N_2$ 0.1       0.6       0.1       0.1 $N_2$ 0.1       0.6       0.1       0.1 $N_2$ 0.1       0.6       0.1       0.2 $N_2$ 0.1       0.2       0.2       0.2 $N_2$ 0.2       0.2       0.2       0.2       0.2 $N_2$ 0.0       0.0       0.454       1482       3.5       5.0       Am $N_2$	CH,	2.1	20.2	9.4		,27.2	8.4	82.3
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	້ ບໍ	1.2	16.0	0.7			5.2	5.2
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$	7 0	0.7	10.5			4.0	4.6	1.8
$ \begin{array}{cccccccccccccccccccccccccccccccccccc$		9.0	5.2				10.0	1.0
4.7       20.2       39.1       32.7       52.4         63.8       0.6       1.2       4.1         0.1       0.6       0.1       0.1         2.5       3.4       17.3       57.4       8.5         0.2       2.8       0.2         0.       60       454       1482       35       50       An         1a       Atm       300       15       Atm       22       93		9.0		0.7		_	4.4	0.8
63.8       0.6       1.2       4.1         0.1       0.6       0.1         2.5       3.4       17.3       57.4       8.5         0.2       2.8       0.2         0       60       454       1482       35       50       An         1a       Atm       300       15       Atm       22       93	о <u>ж</u>	4.7	20.2	39.1	32.7	52.4		
0.1       0.6       0.1         2.5       3.4       17.3       57.4       8.5         2.8       0.2         oc       60       454       1482       35       50         ta       Atm       300       15       Atm       22	V K	63.8		9.0	1.2	4.1		4.8
2.5     3.4     17.3     57.4     8.5       2.8     0.2       oc     60     454     1482     35     50       ta     Atm     300     15     Atm     22	7 0	0.1		9.0		0.1		
oc     60     454     1482     35     50       ia     Atm     Atm     300     15     Atm     22	7 8	2.5	3.4	17.3	57.4	8.5		
oC         60         654         1482         35         50           ia         Atm         Atm         300         15         Atm         22	HCN			2.8		0.2		
ia Atm Atm 300 15 Atm 22		09	. 09	454	1482	35	50	Amb
	Pressure, psia	Atm	Atm	300	15	Atm	22	939

ase Table III-10.

bsee rable III-19. csee ref 1. dsee ref 2. csee ref 3.

Table 23... Range of Gas Compositions from Direct-Fired Retorts a

	1
Oil assay	62.5 to 125 L/tonne
Gas ratio	224 to 405 sm <sup>3</sup> /tonne
Composition (dry basis)	Ì
Carbon dioxide (CO <sub>2</sub> )	23 to 39 vol %
Hydrogen sulfide (H <sub>2</sub> S)	0.07 to 0.24 vol %
Carbonyl sulfide (COS)	0 to 40 ppm
Ammonia (NH <sub>3</sub> ) <sup>b</sup>	0.6 to 0.7 vol %
Water (H <sub>2</sub> O) (wet basis)	13.5 to 15 vol %
Gas mole ratio	
co <sub>2</sub> /H <sub>2</sub> s	76:1 to 165:1 <sup>c</sup>
NH <sub>3</sub> /H <sub>2</sub> S	2.3:1 to 2.4:1 <sup>đ</sup>
Temperature	57 to 77°C
Higher heating value (dry gas)	$1.52 \times 10^6$ to $4.13 \times 10^6$ j

Based on Paraho PON, Occidental Oil Shale (Cxy) retorts 7 and 8, and Geokinetics retort 18.

bRatio for Geokinetics is 0.07:1.

Chigh ratio occurs at start of Oxy burn and averages about 330:1 during the startup phase. At steady-state conditions an average ratio of about 165:1 is obtained.

dRatio reduces to 2.4:1 at steady state for the Oxy retort.

Table 24. Range of Gas Compositions from Indirect-Heated Retorts

Oil assay	158 L/tonne
Gas ratio	29.9 to 38.5 sm <sup>3</sup> /tonne
Composition (dry basis)	
Carbon dioxide (CO <sub>2</sub> )	16.6 to 20.4 vol %
Hydrogen sulfide (H <sub>2</sub> S)	3.8 to 4.1 vol %
Carbonyl sulfide (COS)	135 to 550 ppm
Carbon disulfide (CS <sub>2</sub> )	0 to 20 ppm
Alkyl mercaptans (RSH)	35 to 165 ppm
Sulfur dioxide (SO <sub>2</sub> )	0 to 125 ppm
Ammonia (NH <sub>3</sub> ) <sup>b</sup>	8 vol % <sup>C</sup>
Water (H <sub>2</sub> O) (wet basis)	20 to 25 vol %
Gas mole ratio	• :
∞ <sub>2</sub> /н <sub>2</sub> s	4.3:1 to 5.0:1
NH <sub>3</sub> /H <sub>2</sub> S	2.1
Temperature	60 to 71.1 °C
Higher heating value (dry gas)	$35.20 \times 10^6$ to 42.46 joules/su

<sup>&</sup>lt;sup>a</sup>Based on Tosco-II and Union SGR-3 retorts.

bAbsorbed in sour water condensed with the oil.

<sup>&</sup>lt;sup>C</sup>Calculated value.

## Table 25. Selectivity Data

Selectivity =

$$\frac{\text{(H}_2S \text{ feed - H}_2S \text{ treated gas)/H}_2S \text{ feed}}{\text{(CO}_2 \text{ feed - CO}_2 \text{ treated gas)/CO}_2 \text{ feed}} = \frac{\text{H}_2S \text{ Claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ Claus gas/CO}_2 \text{ feed}} = \frac{\text{H}_2S \text{ Claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ Claus gas/CO}_2 \text{ feed}} = \frac{\text{H}_2S \text{ Claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ Claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ Claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ Claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ Claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ Claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ Claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ Claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ Claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ Claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ Claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ Claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ Claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ Claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ Claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ Claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ claus gas/H}_2S \text{ feed}} = \frac{\text{H}_2S \text{ claus gas/H}_2S \text{ feed}}{\text{CO}_2 \text{ clau$$

$$= \frac{\text{CO}_2 \text{ feed}}{\text{H}_2 \text{S feed}} \times \frac{\text{H}_2 \text{S Claus gas}}{\text{CO}_2 \text{ Claus gas}} = \frac{\text{mole % H}_2 \text{S absorbed}}{\text{mole % CO}_2 \text{ absorbed}}$$

For direct-fired retorts

$$\frac{\text{CO}_2 \text{ feed}}{\text{H}_2 \text{S feed}} = \frac{76}{1} \text{ to } \frac{165}{1}$$

For indirect-fired retorts

$$\frac{\text{CO}_2 \text{ feed}}{\text{H}_2 \text{S feed}} = \frac{5}{1}$$

For Claus feed gas

$$\frac{\text{H}_2\text{S}}{\text{CO}_2}$$
 >0.08 to 0.25

Selectivity required

Direct-fired retorts = 
$$\frac{.76}{1}$$
 X (0.08 to 0.25) = 6 to 19  
=  $\frac{.165}{1}$  X (0.08 to 0.25) = 13 to 41  
Indirect-fired retorts =  $\frac{.5}{1}$  X (0.08 to 0.25) = 0.4 to 1.25

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Claus sulfur recovery process, the acid-gas feed to the Claus unit should be as rich in H2S as possible and consist of at least 8 to 25%  $\rm H_2S$ . Gases from direct-fired retorts with  $\rm CO_2/H_2S$  ratios in the range of 100:1 to 200:1 would require H<sub>2</sub>S selectivity in the range of 8 to 50 to produce an acceptable Claus gas (see Table 25 ); however, gas from indirect-fired retorts would produce an acceptable Claus gas only if all the CO2 were removed along with the H2S.

The quantity of H2S and NH3 in the gas per ton of shale processed is, 26 about equal for both direct- and indirectas shown in Table retorting processes. The quantity of sulfur contained in an equal volume of gas, on the other hand, is as much as 60 times greater for indirect-fired retorts than for direct-fired retorts. Thus a gas treatment system for a direct-fired retort would have to process up to 60 times more gas per ton of sulfur recovered than would a system processing gas from indirect-heated retorts.

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#### B. COMMERCIAL OIL-SHALE OPERATIONS

The projected sizes of several commercial oil-shale processing facilities are given in Table 27. All full-scale processing facilities will consist of multiple retort systems with several trains of qas treating equipment.

The data given in Tables 22 and 23 are based on the gas discharged from the retort after the oil is separated. The actual composition and condition of the gas as it reaches the sulfur removal train may be somewhat different for a full-scale commercial plant if additional product recovery and gas treating steps are incorporated. For most direct-fired-retort processes water scrubbing will be used to remove ammonia from the gas and to cool the gas before it enters the sulfur removal equipment. Some developers propose to compress and then cool the gas to recover additional hydrocarbons. Some may elect to upgrade the raw crude by coking and hydrotreating and/or direct hydrogenation. Residual gases from these operations would likely be added to the feed to the sulfur removal units.

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Table 26. Comparison of Gas Treatment Duty Requirements

	Direct-Fired Retorts	Indirect-Heated Retorts
Duty Range Per 45,4	50 Tonnes of Shale Proces	ssed
Gas rate (Msm <sup>3</sup> D)	10.2 18.4	1.41.8
Hydrogen sulfide (tonnes)	33.6-64.5	78.285.4
Other sulfur compounds (tonnes)	0.042.0	0.63.3.
Carbon dioxide (tonnes)	4,54532,087	443545
Ammonia (tonnes)	3895	83106
Total sulfur (tonnes)	3262	7483
Duty Range Per Milli	on Cubic Feet of Gas Tr	eated
Shale processed (tonnes)	70126	734-947
Hydrogen sulfide (kg)	30103	1,6361,773
Other sulfur compounds (kg)	.53	1264
Carbon dioxide (kg)	12,72721,818	9,25011,363
Ammonia (kg)	130151	1,7271,818
Total sulfur (kg)	2999:	1,5471,710

Table 27. Commercial Oil-Shale Operations

				4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4 4				* ;;;;;;	1		
1			Near-Term D	Near-Term Development Capacity	pacity	7.7		Shale			Sulfur
			Shale	•	•	sul fur	No	Processed	Crude Oil	Raw Gas	in Gas
Type	, Z	No. of Reforts	Processed (tonnes/	Crude 011 Produced	Raw Gas Produced	tonnes/	Retorts	(tonnes/	Produced	Produced	(tonnes/
Company Process		Operating	day)	(m <sup>3</sup> /day)	(PEms M)	day)	Operating	day)	(m-/ day)	(D_1118 13)	(day)
Total	10.				90.	0.0	16	5,454	320	2.8	
	,	ı		4	ć	c	7	143.094	15.867	32.1	120
Paraho <sup>b</sup> Direct	ect.		10,000	960	2.23	ъ	<b>+</b> 7	100			
	4	·	5,363	336	1.58	5.5	40	149,076	9,120	48.3	131
Occidental Direct	נ נ						•	000	008 8	6	175
Tosco II d Indi	Indirect	9	000,09	8,800	2.32	174.5	٥	90,000	999	) :	•
	Indirect	-	060'6	1,440	0.27	14.5	9,	47,886	8,000	1.4	120

ase refs 4 and 5.

bsee refs 5, 6 and 7.

csee refs 7 and 8.

dSee refs 7 and 9.

eSee refs 10 and 11.

Indirect-retort gases that have high heating values may be combined with residual gases from other process steps and may undergo several treatment steps before they reach the sulfur removal equipment. For instance the shale-oil hydrocarbon vapors from the pyrolysis drum of the proposed Colony Tosco-II facility will be separated in a fractionator into sour water, gas, naphtha, gas oil, and bottom oil. The gas and naphtha are piped to a gas recovery and treating unit. The other streams are further treated to upgrade the product. The residual gases from product upgrading units are combined with the gas stream from the fractionator and then compressed and fed to the gas recovery and treating unit along with raw naphtha recovered from various treatment units. Stabilized naphtha, LPG, butanes, butadiene/butylenes, and ammonia are separated and recovered. The remaining gas then enters the sulfur removal equipment. Acid gas with residual ammonia from the sour-water stripper and from the ammonia separation unit is sent directly to the

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Claus sulfur recovery unit.

For the purpose of this report it is assumed that the composition of the gas is as discharged from the retort after the oil has been separated.

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- 1. M. Gmassemi et al., TRW, Inc., Applicability of Petroleum Refinery Control Technologies to Coal Conversion, EPA-600/7-78-190 (October 1978).
- 2. H. Hiraoka, Mitsubishi Chemical Industries, Ltd., and E. Tanaka and H. Sudo, Mitsubishi Kakoki Kaisha, Ltd., "DIAMOX Process for the Removal of H<sub>2</sub>S in Coke Oven Gas," Proceedings of The Symposium on Treatment of Coke-Oven Gas, May 1977, McMaster University Press, Hamilton, Ontario, Canada.

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3. W. L. Scheirman, "Sour Gas Treating at the World's Largest Natural Gas Processing Plant," p. 0-8 in <u>Proceedings of the Gas Conditioning Conference</u>, <u>March 7—9, 1977</u>, sponsored by Continuing Engineering Education, University of Oklahoma, Norman.

\*When a reference number is used at the end of a paragraph or on a heading, it usually refers to the entire paragraph or material under the heading. When, however, an additional reference is required for only a certain portion of the paragraph or captioned material, the earlier reference number may not apply to that particular portion.

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### VII. SCREENING OF GAS TREATING PROCESSES

Gases from direct-fired retorts are more limiting in the application of sulfur-removal technology than are other gases (see Sect. VI-A). In the course of this study it was determined by the EPA that the greatest immediate concern was the control of sulfur emissions from direct-fired processes and that the pilot-plant design should be applicable to these retorting methods. Therefore process selection was directed toward the requirements for desulfurization of gas from direct-fired retorts.

As is indicated in Table 23, treating gas from direct-fired retorts requires a process with high selectivity toward  $H_2S$ . The initial screening of principal commercial gas-conditioning processes was therefore based on the ability of each process to selectively remove  $H_2S$  in the presence of large amounts of  $CO_2$ . Those processes most capable of selectively removing  $H_2S$  were selected as candidates for more detailed evaluations. The processes referred to are shown on Fig. 9, a duplicate of Fig. 6. For the purpose of the initial screening, selectivities near the midpoint of the selectivity range reported in the literature were used (see Table 28 ).

# A. DIRECT-CONVERSION PROCESSES

In the direct-conversion processes the sulfur compounds are directly oxidized to elemental sulfur or are converted to another compound, which is then separated or recovered.

# 1. Dry Bed

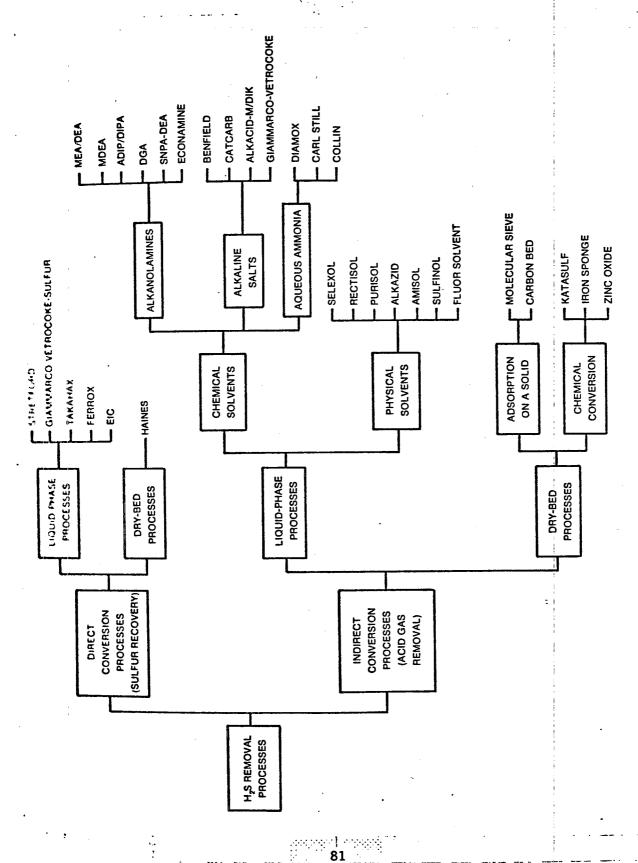
Several dry-bed direct-conversion processes have been tested but none are known to have been commercially applied. These processes are similar to the Claus process except that the SO<sub>2</sub> necessary for the reaction is obtained by burning sulfur in an external burner and the gas to be treated is preheated to the operating temperature of the catalyst beds. To maintain high conversion efficiency the catalyst beds are regenerated more frequently than is required by the Claus process. One version is designed so that the catalyst is continually

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Gas-Sweetening Processes Figure 9.

Table 28. Selectivity of Absorbent Processes

Absorbent	Selectivity Range	Reported By
MEA	0.89	Pearce <sup>a</sup>
DEA	1.2 to 2.27	Pearce
MDEA	1.2 to 7.0	Pearce
DIPA	3.33 to 5	Naber et al.b
NH <sup>3</sup>	1.00 to 28.9	Kohl and Riesenfeld
Diamox	5.7 to 9.4	Hiraoka <u>et</u> al. d
Benfield	2.25 to 9.8	Parrish and Neilson
Selexol	5 to 22	Kohl and Riesenfeld

a<sub>See</sub> ref 5.

b<sub>See</sub> ref 6.

c<sub>See ref 10.</sub>

d See ref 13.

e<sub>See</sub> ref 9.

fSee ref 14.

withdrawn from the bottom of the bed and the regenerated catalyst is --- added at the top. 1

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The Haines process uses a bed of synthetic zeolite to adsorb the  $H_2S$ . The bed is then regenerated with high-temperature gas that contains sulfur dioxide. The sulfur formed is condensed and recovered.

The application of these processes is limited due to the possibility that the beds will be plugged by condensible hydrocarbons and/or other impurities in the gas. Since these processes have not been commercially proven, they are not considered further.

# 2. Liquid Phase

Except for the EIC process the principal liquid-phase direct-conversion processes shown on Fig. 9 selectively remove H<sub>2</sub>S by converting it directly to elemental sulfur. Carbon dioxide is only slightly absorbed and largely remains in the treated gas. With the EIC process, sulfur is removed as ammonium sulfate and CO<sub>2</sub> is not removed.

The Stretford, Giammarco-Vetracoke (G-V), and Takahax processes are based on similar oxidation-reduction chemistry but use different activator chemicals. The G-V process uses an arsenic compound, but because of arsenic's toxicity the process is not applied in this country. The Takahax process, developed in Japan, has not been widely applied.

The Ferrox process, which is based on the reaction between iron oxide and H<sub>2</sub>S, is outdated and has been replaced by the more efficient Stretford Process. In the Stretford process, the direct-conversion process most often employed, H<sub>2</sub>S is selectively removed by direct oxidation to elemental sulfur but CO<sub>2</sub> is only slightly absorbed. The process is usually the preferred choice when the H<sub>2</sub>S concentration in the raw gas is very low or when a high degree of selectivity is required. Complete removal of H<sub>2</sub>S to about 1 ppm can be obtained in the treated gas. Since the sulfur absorption capacity of the Stretford solution is low, the process is not competitive with indirect sulfur removal processes

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for volumes of  $\rm H_2S$  in the feed gas that exceed about 16% of the total acid-gas content of the gas.<sup>2</sup>

Organic sulfur compounds are not removed by the Stretford process but remain in the treated gas. Stable thiosulfates form in the oxidizer equivalent to about 1% of the sulfur in the feed gas. A portion of the circulating solution must be purged to control the buildup of these compounds. The quality of sulfur produced is low compared to that produced by the indirect-recovery processes. In spite of its shortcomings the Stretford process has been widely applied, and when the CO<sub>2</sub>/H<sub>2</sub>S ratio is very high, it may be the only process available that is capable of economically purifying the gas.

The EIC process is an absorption-oxidation process using copper sulfate as the absorbent that produces by-product ammonium sulfate from the hydrogen sulfide and ammonia in the treated gas. The process reportedly removes H<sub>2</sub>S and COS and ammonia but does not remove CO<sub>2</sub>. A high removal efficiency of 99% is obtained in one absorption step and the system can operate at elevated temperature. The process is in the development stage and has not been commercially proven. However, since it appears to match the requirements needed for desulfurization of direct-fired oil-shale gases, it will be considered as a canidate for further evaluation.

# B. INDIRECT-CONVERSION PROCESSES

In the indirect conversion processes the acid-gas components are removed from the fired gas and recovered as a separated stream that is subsequently processed for recovery of sulfur.

1. Dry Bed

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The dry-bed indirect-conversion processes provide excellent selectivity, with most of them capable of removing sulfur compounds with little or no CO<sub>2</sub> removal. These processes use a fixed bed of solid material. Processes that use molecular sieves or carbon adsorb the sulfur compounds, which are subsequently released unchanged during regeneration

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> of the bed. With the iron sponge and zinc oxide processes,  ${\rm H_2S}$  reacts with the bed to form iron or zinc sulfide. The Katasulf process is based on the catalytic oxidation of  $H_2S$  to  $SO_2$ , followed by removal of  ${\bf SO}_2$  with an ammonium sulfite—bisulfite solution.

> Dry-bed processes, however, are limited in their application and become economically impractical for those cases where large amounts of gas have to be treated or the total quantity of sulfur to be removed is high. Since full-scale oil-shale facilities will be processing huge amounts of gas beyond the practical limit for application of dry-bed processes, these processes are not considered further.

# Liquid Phase

With liquid-phase indirect-conversion processes the acid-gas components are removed from the feed gas by absorption into liquid. The liquid is subsequently regenerated to produce a concentrated stream of acid gas, which is then processed by the Claus process for recovery of the sul-The principal liquid-phase indirect-conversion processes are shown on Fig. VII-1 and are classified by type.

Alkanolamines -- The alkanolamine processes are based on the reaction of a. a weak base (alkanolamine) and a weak acid ( ${\rm H}_2{\rm S}$ , organic acids, and/or CO2) to form a water-soluble salt:

$$RNH_2 + H_2S \longrightarrow RNH_3^+ \cdot HS$$
 $RNH_2 + CO_2 + H_2O \longrightarrow RNH_3^+ \cdot HCO_3^-$ 

These reactions are reversible, and the equilibrium may be shifted by adjustment of the solution temperature. The absorption of CO2 involves the formation of carbamates as intermediate compounds. With primary and secondary amines the carbamates are formed nearly instantly and the reaction rate of  ${\rm CO_2}$  with the amine is nearly equal to that of  ${\rm H_2S}$  with amine. With tertiary amines carbanates form at a much slower rate and therefore H2S is absorbed more readly than is CO2. This difference in FORTABLES

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reaction rates between  $H_2S$  and  $CO_2$  accounts for the higher selectivities that are obtained with tertiary amines. Methyldiethanolamine (MDEA) reportedly produces the highest selectivity of the tertiary amines<sup>4</sup>—6 (see Table 28).

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The selectivity at which  ${\rm H_2S}$  can be absorbed in the presence of  ${\rm CO_2}$ depends on absorption kinetics rather than on equilibrium effects. The actual selectivity obtained by a process is dependent on several variables, the most readily controlled being the absorber contact time. In practice high selectivity is obtained by limiting the number of contact stages in the absorber and the residence time per stage. Absorption kinetics are also affected by a variety of other process variables, including competition between H2S and CO2 when they are absorbed simultaneously. The degree of H2S removal decreases as the contact time is reduced in an attempt to increase the selectivity. The extent of selectivity that can be obtained will be restricted by the purity required of the treated gas. To reasonably predict these effects for a specific gas composition and at a specified degree of desulfurization, the process developers have devised proprietary computer programs for modeling the absorption process. 7'8 These models were developed and . verified through extensive laboratory and field testing programs.

The average selectivity of MDEA as reported by Pearce<sup>5</sup> is 3.85. If a two-stage selective absorption system were used and a selectivity of 3.85 per stage were obtained, the H<sub>2</sub>S in the acid-gas stream would be about 12.9 vol % for a feed gas with a CO<sub>2</sub>/H<sub>2</sub>S ratio of 100:1 (see Table 29). If the ratio were 200:1, the acid gas would contain about 6.9% H<sub>2</sub>S, which is marginally acceptable. A possible process option would be a system consisting of two or more stages of selective absorption, with MDEA used as the solvent.

Alkaline Salts—The principal alkaline salt processes, shown on Fig. VII-1, are most successfully applied for bulk CO<sub>2</sub> removal but are not generally considered when selective absorption of H<sub>2</sub>S is required.

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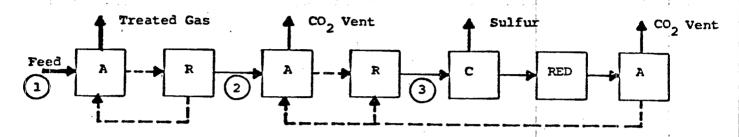
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Table 29 Selective Absorption Using MDEA with Two Stages of Absorption



A = absorption; R = regeneration; C = Claus; RED = reduction.

**************************************	CO <sub>2</sub> /H <sub>2</sub> S	
Stream	Ratio	H <sub>2</sub> S (vol %) <sup>b</sup>
Case 1	•	• !
1 Feed 1st stage	100	1.0
② Feed 2nd stage	26	3.7
3 Claus gas	6.7	12.9
Case 2		
1 Feed 1st stage	200	0.5
Peed 2nd stage	51.9	1.9
3 Claus gas	13.5	6.9
Case 3		<u> </u>
<ol> <li>Feed 1st stage</li> </ol>	4	20.0
2 Feed 2nd stage	1.04	49.0
3 Claus gas	0.27	78.7

Based on obtaining selectivity ratio of 3.85 per stage.

bAs vol & of total CO2 and H2S.

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The Catacarb process is not highly selective in  ${\rm H}_2{\rm S}$  removal and the Giammarco-Vetrocoke process, designed for bulk CO2 removal, uses an arsenic compound to improve the solution reactivity. The Alkacid process using DIK solvent will selectively absorb  $\mathrm{H}_2\mathrm{S}$ . However, the capacity of the solution for absorbing  $H_2S$  and the degree of  $H_2S$  removal obtained decrease as the the  ${\rm CO_2/H_2S}$  ratio of the feed gas increases.

The Benfield process is normally applied for bulk removal of acid gas. A high degree of  ${\rm H_2S}$  removal can be obtained when the  ${\rm CO_2/H_2S}$  ratio is high and both acid gases are simultanelusly absorbed. However, the acid-gas stream produced is usually too lean for Claus processing. Since  $\mathrm{H}_2\mathrm{S}$  is absorbed more rapidly than  $\mathrm{CO}_2$  in a potassium carbonate solution, the process can be made to selectively absorb  $H_2S$ ; but, as the selectivity increases, the degree of removal of  ${
m H}_2{
m S}$  decreases.

The Benfield HiPure process, a two-stage process using an amine solution in the second stage to absorb residual  $\mathrm{H}_2\mathrm{S}$  and optimized to take advantage of the higher solubility and absorption rate of  $\mathrm{H}_2\mathrm{S}$ , has produced selectivities of up to 9.8 based on data reported by Parrish and Neilson. 9 If an average selectivity of 6 were obtained with the Benfield HiPure process (see Table 30), the H<sub>2</sub>S content of the acid gas would be approximately 5.7 for a feed gas with a  ${\rm CO_2/H_2O}$  ratio of 100:1 or 2.9 if the ratio were 200:1. In either case insufficient selectivity would be obtained to provide an acceptable Claus gas.

To apply the Benfield process, the feed gas must be at high pressure, above 100 psig. The higher the pressure, the more efficient the process and the higher the selectivity that is obtained. Since gas from proposed oil-shale retorts would be produced at near-atmosphere pressure, the gas would have to be compressed. The capital and operating costs of compressing the gas cannot be justified by use of the Benfield process. Also, since  $H_2S$  is less soluble in the alkaline salt solution at higher temperatures, the advantage that hot gases can be fed to the process is offset somewhat in that selectivity decreases as the temper-

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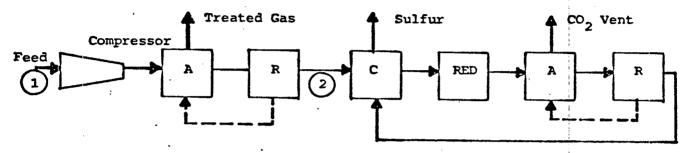
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Table 30. Selective Absorption by Benfield Process



A = absorption; R = regeneration; C = Claus, RED = reduction.

Ctrops.	CO <sub>2</sub> /H <sub>2</sub> S Ratio	, b
Stream	Ratio	H <sub>2</sub> S (vol %)
Case 1	••	<u>.</u>
① Feed	100	1.0
② Claus gas	• 17	5.7
Case 2		
1 Feed	200	0.5
2 Claus gas	33	2.9
Case 3		1
1 Feed	4	20.0
2 Claus gas	0.7	60.0

Based on selectivity ratio of 6.

bAs vol % of total CO<sub>2</sub> and H<sub>2</sub>S.

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Aqueous Ammonia Processes Aqueous ammonia processes have been used since the end of the nineteenth century for removal of hydrogen sulfide and nitrogen compounds, primarily ammonia from coal gas. Many ammoniabased processes have been developed over the years; the principal ones used today are shown on Fig. 9. Ammonia-based processes are particularly applicable to gases containing both  ${
m H}_2{
m S}$  and ammonia. ammonia contained in the gas can serve as the active agent for removal of H<sub>2</sub>S and can be recovered as a by-product.

If the physical parameters of the absorption process are controlled, aqueous ammonia solution can selectively remove H2S from gases containing CO<sub>2</sub>. Selectivities of up to 28.9 have been reported. 10 Hydrogen sulfide ionizes immediately on contact with the solution and reacts rapidly with the hydroxyl ions, whereas carbon dioxide must first react with water, forming carbonic acid, before it can react ionically with ammonia. The rate of the CO<sub>2</sub> hydration reaction is quite low compared to the reaction rate of  $\rm H_2S.^{11}$ 

The absorption of ammonia into water is quite rapid and is governed principally by the gas-film resistance. The rate of absorption of  ${\rm H}_2{\rm S}$ into aqueous ammonia solutions is also rapid although it is dependent on the ammonia concentration and is probably also governed by the gasfilm resistance. Carbon dioxide absorption into water or weak alkaline solutions, on the other hand, is governed by its liquid-film resistance, which is very much greater than that for  $H_2S$ . As a result when gases containing  $\mathrm{H}_2\mathrm{S}$ , ammonia, and  $\mathrm{CO}^2$  are contacted with water or ammonia solution, the ammonia and H<sub>2</sub>S are absorbed much more rapidly than is CO2. This difference can be accentuated by operating under conditions that reduce the gas-film resistance or increase the liquidfilm resistance. Thus to achieve maximum selectivity, spray columns in combination with relatively short contact times are used. 11 Table 31 shows the results that could be obtained if a selectivity of 28.9 were achieved. However, the degree of H<sub>2</sub>S removal decreases as the selectivity is increased. Removal of about 90% of the  ${\rm H_2S}$  is the maximum efficiency that can be attained economically through selective ab-AND ILLUS 3./8\*\*

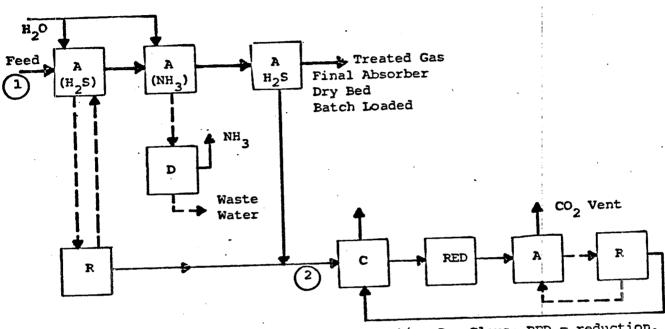
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Table 31. Selective Absorption Using Aqueous Ammonia



A = absorption; D = distillation; R = regeneration; C = Claus; RED = reduction.

Stream	CO <sub>2</sub> H <sub>2</sub> S Ratio <sup>a</sup>	H <sub>2</sub> S (Vol %)
		·
Case 1	***	1
1 Feed	100	22.4
2 Claus gas	3.46	· · · · · · · · · · · · · · · · · · ·
Case 2		0.5
1 Feed	200	12.6
2 Claus gas	6.92	12.0
Case 3		20.0
1 Feed	4	87.8
2 Claus gas	0.14	87.0

Based on selectivity ratio of 28.9.

bAs vol & of total CO2 and H2S.

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sorption. A final purification step must be added to remove the residual  $H_2S$  if it is desired to achieve a high degree of removal with a high degree of selective absorption.

The selectivity of aqueous ammonia solutions decreases markedly at temperatures above 27°C.12 Also, the selectivity decreases as the ammonia concentration of the solution is increased. However, the degree of H2S removal increases as excess available ammonia is increased. The Diamox process, recently developed in Japan, was designed to take maximum advantage of these factors. Hydrogen sulfide removal in the range of 97 to 99%, with good selectivity still maintained, can be achieved by the Diamox process. Selectivities in the order of 5.7 to 9.4 have been obtained based on data reported by Hiraoka, Tanaka, and Sudo. 13 If an average selectivity of 7.6 were obtained with the 32), an acid gas containing 7.1% would be ob-Diamox process (Table tained for a feed gas with a CO<sub>2</sub>/H<sub>2</sub>S ratio of 100:1, or 3.7% H<sub>2</sub>S would be obtained if the ratio were 200:1. The Diamox process would be marginally acceptable for gases with CO2/H2S ratios of up to about 100:1 but would not produce sufficient selectivity for direct-fired retort gases with higher CO2/H2S ratios.

d. <u>Physical-Solvent Processes</u>—Selectivity of physical-solvent processes depends on the relative solubilities of CO<sub>2</sub> and H<sub>2</sub>S in the solvent. The capacity of the solution for absorbing H<sub>2</sub>S increases with increasing pressure and decreasing temperature; generally the higher the partial pressure of H<sub>2</sub>S the higher the selectivity attainable will be. Selectivity can also be enhanced by favorable absorber kinetics.

The Selexol process appears to produce the highest reported selectivity of the physical-solvent processes;  $H_2S$  is 9 times more soluble in the Selexol solvent than  $CO_2$  is. Selectivity ratios of up to 22 have been obtained based on data reported by Kohl and Riesenfeld.<sup>14</sup>

To apply the Selexol process the retort gas would have to be compressed and cooled before it entered the absorber. If an average selectivity

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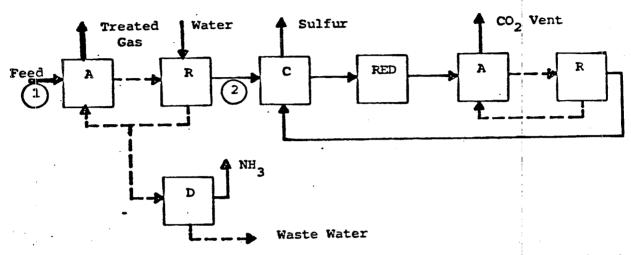
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Table 32. Selective Absorption by Diamox Process



A = adsorption; D = distillation; R = regeneration; C = Claus; R = reduction.

•	•	
Stream	CO2/H2S Ratio	H <sub>2</sub> S (vol %) <sup>b</sup>
Case 1		
1 Feed	100	1.0
② Claus gas	13.2	7.1
Case 2		:
1 Feed	200	0.5
② Claus gas	26.3	3.7
Case 3	•	
(1) Feed	4	20.0
2 Claus gas	0.53	65.5

Based on selectivity ratio of 7.6.

bas vol & of total CO2 and H2S.

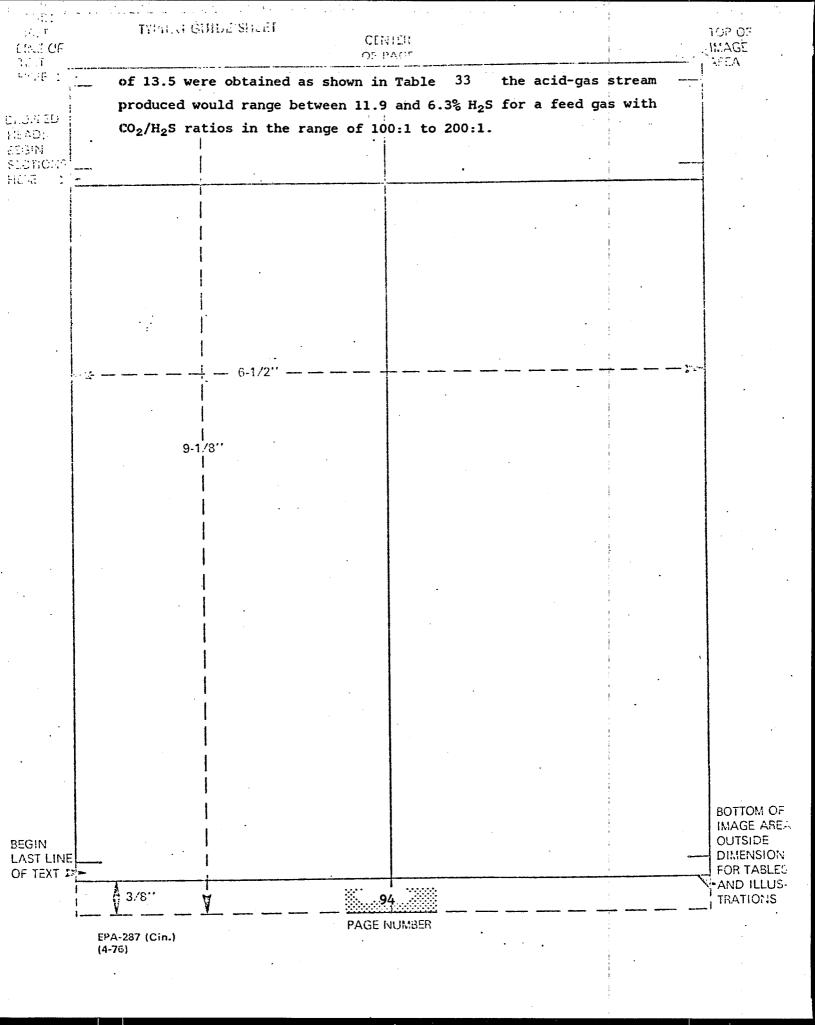
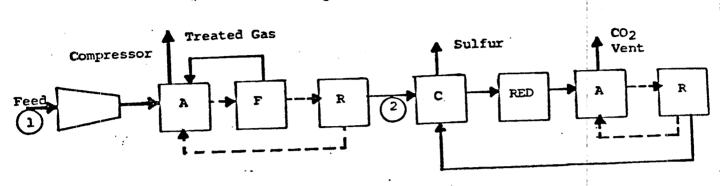


Table 33. Selective Absorption by Selexol Process



A - absorption; F = flash; R = regeneration; C = Claus; RED = reduction.

	CO <sub>2</sub> /H <sub>2</sub> S	b
St. wo nm	Ratio	H2S (vol %)
Stream	•	• •
Case 1		1.0
1 Feed	100	
② Claus gas	7.4	11.9
_		
Case 2	200	0.5
1 Feed		6.3
2 Claus gas	14.8	
Case 3		20.0
(1) Feed	4	1
$\simeq$	0.30	77.1
② Claus gas		i i

Based on selectivity ratio of 13.5.

bAs vol & of total CO2 and H2S.

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- A. L. Kohl and F. C. Riesenfeld, op. cit., p 148. 10.
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- 12. Ibid., p 151.
- H. Hiraoka, Mitsubishi Chemical Industries, Ltd., and E. Tanaka and H. Sudo, Mitsubishi Kakoki Kaisha, Ltd., "DIAMOX Process for the Removal of H<sub>2</sub>S in Coke Oven Gas," Proceedings of the Symposium on Treatment of Coke-Oven Gas, May 1977, McMaster University Press, Hamilton, Ontario, Canada.
- 14. A. L. Kohl and F. C. Riesenfeld, op. cit., pp 784-789.

\*When a reference number is used at the end of a paragraph or on a heading, it usually refers to the entire paragraph or material under the heading. When, however, an additional reference is required for only a certain portion of the paragraph or captioned material, the earlier reference number may not apply to that particular portion.

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# VIII. EVALUATION OF CANDIDATE PROCESSES

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As determined from the data given in Sect. VII the processes that produce the highest H<sub>2</sub>S selectivity or appear to be most applicable to treatment of gases from direct-fired oil-shale retorts are the following:

# Direct-Conversion Processes

Stretford

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# Indirect-Conversion Processes

Alkanolamine processes—MDEA

Alkanline salt processes—Benfield

Aqueous ammonia processes—Diamox

Physical solvent processes—Selexol

# A. BASIS OF EVALUATION

To furthur evaluate these processes their capabilities were compared, with a hypothetical feed gas composition used as the basis for comparison. The gas composition chosen (Table 34) is similar to the gas produced by the Paraho direct-fired retort. Trace organic sulfur compounds, not included in Paraho's data, were added so that their effects could be evaluated. The quantity of trace sulfur compounds shown was extrapolated from data compiled from sources reporting on trace compounds. 1

The indirect sulfur removal processes are only marginally capable of producing an acceptable Claus gas when the  $\rm CO_2/H_2S$  ratio of the feed gas exceeds about 100:1 (see Tables 29 through 33). The hypothetical gas, with a  $\rm CO_2/H_2S$  ratio of 73:1, was chosen so that it would be within the capabilities of the more marginal processes. To treat this gas, using an indirect removal process, an  $\rm H_2S$  selectivity ratio of at least 6 would be required to produce a minimally acceptable acid gas for the Claus unit.

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Table 34. Hypothetical Direct-Fired Retort Gas

Physical Properties	Product Gas Composition (wet basis)	Amount (wet vol %)
Retort off-gas temperature: 66°C		
Retort off-gas pressure: 20 psia		;
Gas rate: 224 sm <sup>3</sup> /tonne of shale		
Gas density: 1.2 kg/m <sup>3</sup>		
Molecular weight: 28.43		1
	H <sub>2</sub>	3.539
•	02	0.743
	N <sub>2</sub> + Ar	51.342
	ထိ	1.600
	co <sub>2</sub>	17.572
	CH <sub>4</sub>	1.858
	C2H4	0.743
·	C <sub>2</sub> H <sub>6</sub>	0.768
	С <sub>3</sub> н <sub>6</sub>	0.364
	C <sub>3</sub> H <sub>8</sub>	0.380
	$C_4^{S}$ (MW = 57.0)	0.323
	C <sub>5</sub> S (MW = 71.5)	0.129
	$C_6 + (MW = 96.2)$	0.622
	H <sub>2</sub> S	0.242
•	NH <sub>3</sub>	0.566
	н <sub>2</sub> о	19.210
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aIncludes the following trace compounds: COS, 36 ppm; CS<sub>2</sub>, 7 ppm; RSH, 10 ppm; and SO<sub>2</sub>, nil.

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Projected plans for full-scale commercial oil-shale plants based on direct-fired retort technology range from Geokinetics' facility producing 2.83 million m<sup>3</sup> of gas per day from 16 retorts processing about 5454 tonnes of shale per day<sup>2</sup> to Occidental's MIS plant producing 48 million m<sup>3</sup> of gas per day from a series of 40 retorts processing about 149,078 tonnes of shale per day.<sup>3</sup>

The model plant selected for evaluation of a sulfur removal process has a capacity of 45,450 tonnes of shale per day and produces 10.2 million  $\mathfrak{P}^3$  of gas a day based on the composition of the hypothetical gas. A desulfurization plant in this size range will require multiple trains to process the volume of gas produced, as would be the case for most projected commercial oil-shale facilities. The individual components required for the model-plant sulfur removal train should therefore be in the size range of that required for a full-scale commercial oil-shale facility and should be representative of such a facility.

# B. GAS PRETREATMENT

Most sulfur removal processes will require some form of pretreatment of the gas before it enters the sulfur removal equipment. For most processes the gas must be cooled and ammonia, condensible hydrocarbons, and excess water be removed. Other processes may require compression of the gas. Particulate matter is largely removed in the oil separation and gas cooling steps. Additional particulate removal is not required before the gas enters desulfurization equipment.

1. Gas Cooling and Ammonia Removal

The temperature of raw gases produced by direct-fired oil-shale retorts range from 57° to 77°C. For most sulfur removal processes the gas must be cooled before it enters the absorber. Since the gases from direct-fired retorts are saturated with water, large amounts of water will be condensed out as the gas is cooled. Calculations indicate that sufficient water can be condensed to simultaneously absorb and remove the ammonia contained in the gas if the gas is cooled to about 32°C.

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The gas cooling unit for the model plant is shown in Fig. VIII-1. raw fuel gas (stream 1), which is at 60 °C, is directly contacted with the cooling water in a packed tower to bring the temperature down to 32°C. As the gas cools, a large portion of the water vapor, 849 kg/ min, and a small amount of hydrocarbons are condensed into the circulating cooling water. Essentially all the ammonia, a stoichiometric amount of CO2, and a small amount of H2S are absorbed into the cooling water. The cooled gas (stream 3) is sent to the desulfurization system absorber.

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A portion of the cooling water solution equilivent to the quantity of gases and vapors absorbed is purged (stream 2) from the system. remaining cooling water solution, saturated with CO2, is passed through a heat exchanger to remove the heat absorbed and then recirculated. The operating conditions and the composition of the recirculating cooling water are controlled so that minimum H2S is absorbed into the cooling water.

The purge stream will be treated (not shown) by being passed through an oil separator, where condensed hydrocarbons are recovered, and then through a sour-water stripper and ammonia recovery unit. The H2S recovered in the sour-water stripper is returned to the process gas stream. The ammonia is stripped and recovered as a by-product. The remaining water, which contains trace amounts of absorbed organics, H2S, and ammonia, is sent to a conventional wastewater treatment system.

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The heavier hydrocarbons with boiling points above 90°F will condense and be largely removed in the gas cooling unit. However, components in the gas include unsaturated hydrocarbons such as acetylene, ethylene, propylene, butadiene, and butylene. These components will not condense as the gas is cooled. They can, however, polymerize and form tarry substances that can foul the acid-gas absorber, discolor the sulfur, or clog the catalyst beds of a Claus sulfur recovery unit if polymeriza-

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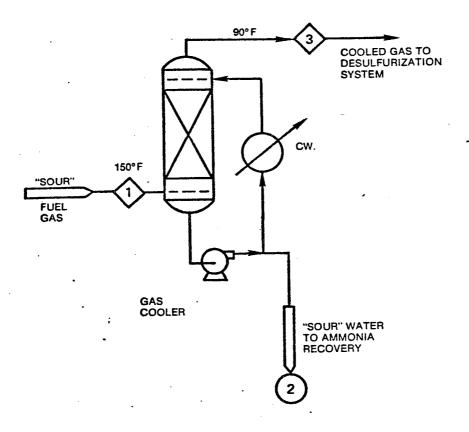


Figure 10. Gas Cooler/Ammonia Absorption System

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tion takes place in the acid-gas absorber. There is no economical way of removing these compounds before the gas enters the acid-gas absorber, and their effects on the desulfurization train can be determined only by testing of the process.

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## 3. Gas Compression

Both the Benfield and the Selexol processes require elevated pressure to operate and thus would require gas compression. For the model plant producing 10.2 million sm<sup>3</sup>d of gas (Table 34) a compressor rated at 84,500 hp would be required to boast this pressure to 150 psig. A gasturbine-driven compressor system of this capacity with intercoolers and condensate separation is estimated to cost \$6 to \$8 million and would consume about \$34,500 of fuel per day, or about 32% of the treated gas produced by the facility.

If the treated gas were combusted in a gas turbine used for power generation, a portion of the available energy in the compressed gas would be recovered as the gas expanded through the turbine. Thus if the gas is compressed, it would be advantageous to remove as little CO<sub>2</sub> as possible in the desulfurization process so that more gas will be available to expand through the turbine. However, after compression the gas would have to be cooled to the operating temperature of the sulfur removal process, and so a large portion of the available energy in the form of heat would be removed before the compressed gas reached the power generation turbine. From an energy standpoint it would be more efficient to compress the gas as it enters the power generation turbine rather than before sulfur removal.

Since the Benfield process can operate at up to 121°C, less heat would have to be removed from the compressed gas than with the Selexol or other processes that operate below 32°C. Approximately 4673.6x10<sup>9</sup> joules of heat per day would have to be extracted to cool the compressed gas to 121°C, at a cost of about \$1750/day for cooling water. To cool the compressed gas to 32°C approximately 10,118x10<sup>9</sup> joules/day must be extracted at a cost of \$3850/day.

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Compressing the gas to 150 psig would reduce its volume by a factor of 6 to 10, depending on the exit temperature. Thus the number or size of absorbers required to handle the gas would be reduced porportionately for those processes in which the absorber size is gas limiting (Selexol, Benfield, and to some extent the alkanolamine processes). For those processes in which the absorber size is liquid limiting (Stretford, Diamox, and to some extent MDEA) the size of the absorbers may not be significantly different if the gas were compressed.

Due to the high capital and operating costs of gas compression it is not economical to compress the gas for the purpose of sulfur removal alone. Some full-scale commercial oil-shale plants employing indirectheated retorts may compress the gas before sulfur removal for the purpose of recovering LPG and other condensible hydrocarbons or for other process reasons. In that case the application of the Benfield or Selexol process may be attractive.

#### DIRECT-RECOVERY PROCESSES C.

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The Stretford Process (See Appendix A) 1.

. The Stretford process was developed in England jointly by the North Western Gas Board and the Clayton Aniline Company, Ltd. 4 The process has been proven commercially over a period of many years in more than 80 plants built around the world. 5 Several modifications and improvements to the basic process have been made by various companies that now license their proprietary versions of the Stretford process. principal licensors in this country are Peabody Process Systems of Stamford, Connecticut, the Pritchard Corporation of Kansas City, Missouri, and the Ralph M. Parsons Company of Pasadena, California.

The chemistry of the Stretford process is relatively complex. idealized reactions illustrate the absorption, oxidation/reduction cycle as follows:6

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$$Na_2CO_3 + H_2S \longrightarrow NaHS + NaHCO_3$$

Reaction 2

$$4\text{NaVO}_3$$
 +  $2\text{NaHS}$  +  $\text{H}_2\text{O}$   $\longrightarrow$   $\text{Na}_2\text{V}_4\text{O}_9$  +  $2\text{S}$  +  $4\text{NaOH}$ 

Reaction 3

$$Na_2V_4O_9 + 2NaOH + H_2O + 2ADA \longrightarrow 4NaVO_3 + 2ADA (reduced)$$

Reaction 4

--- 2ADA (reduced) + 
$$0_2 \rightarrow 2ADA + H_2O$$

A flow diagram of the Stretford process for fuel gas desulfurization is shown in Fig. 11. The sour fuel gas (stream 1) enters the gas cooler, where it is cooled and a large portion of the water and most of the ammonia are removed (stream 2) and sent to foul water treatment. The cooled gas (stream 3) enters the Stretford absorber at about  $32^{\circ}$ C. The feed gas (stream 3) is countercurrently washed with an aqueous solution containing sodium carbonate, sodium metavanadate, and anthroquinone disulfonic acid (ADA). The  $\rm H_2S$  rapidly dissolves and ionizes in the alkaline solution, forming a small amount of sodium hydrosulfide (reaction 1). Hydrosulfide loadings in the solution range from 500 to 1000 ppm; thus relative large amounts of solution must be circulated. The sodium metavanadate in the solution readily reacts with the hydrosulfide to produce elemental sulfur (reaction 2).

Sodium carbonate in the solution provides a pH buffer to prevent rapid pH changes as the acid gases are absorbed. The alkalinity of the Stretford solution causes some of the  $CO_2$  to be absorbed along with the  $H_2S$ . The treated gas (stream 4) with essentially all the  $H_2S$  removed exits from the top of the absorber. The aqueous solution enters the delay tank, with sufficient residence time provided to assure that any

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Figure 11. Stretford Process for Fuel-Gas Sulfurization

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remaining hydrosulfide is reacted so as to prevent hydrosulfide from entering the oxidizer and becoming oxidized to stable sodium thiosulfate, an undesirable by-product. Thiosulfate can be formed by oxygen dissolved in the Stretford solution, oxygen in the feed gas, or hydrosulfide carryover to the oxidizer. An excess of vanadate is maintained to avoid overloading the solution with hydrosulfide. Formation of thiosulfate can be controlled somewhat by controlling the pH and the temperature of the solution. Thiosulfate can be allowed to concentrate in the solution to about 20% by weight. A small portion of the solution (stream 14) is then continually purged and replaced to prevent further accumulation and crystallization of salts.

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The reduced solution (stream 6) is regenerated in the oxidizer tower by being sparged with an excess of air. The reduced vanadium is reoxidized to its original state through oxygen transfer via the ADA (reaction 3). The reduced ADA is reoxidized by contact with air (reaction 4). Reoxidation of the ADA can be appreciably accelerated by the presence of small amounts of iron salts kept in solution by a chelating agent, ethylenediamine tetracetic acid (EDTA).

The nitrogen and the excess oxygen from the air bubble upward through the solution, floating sulfur particles to the top as a froth and stripping the dissolved gases. The froth, which contains about 10% solids, overflows the oxidizer to a slurry tank. A damp cake containing 50 to 60% solids is produced by filtration or centrifugation of the slurry (stream 10). The sulfur cake can be further processed by washing to remove the Stretford chemicals and by drying and melting to produce liquid or solid sulfur (stream 17).

The regenerated solution (stream 5) is relatively free of sulfur and is pumped back to the absorber. In route the solution must normally be cooled to maintain the desired operating temperature, which is usually accomplished in an evaporative cooling tower. To maintain water balance in the system due to the addition of sulfur and/or filter wash water and water produced by the reaction, water is evaporated with the

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treated gas or in the cooling tower. A solution heater is sometimes required for winter operation or to assure sufficient evaporation.

Carbon dioxide is partly absorbed by the alkaline solution, resulting in the formation of sodium bicarbonate and consequent lowering of pH. The solution eventually reaches equilibrium with respect to the concentration of CO2 in the gas stream, after which only relatively small amounts of CO2 are absorbed. Since the vanadate-ADA system functions at a lower pH, decarbonation of the solution is not required. When the gas contains high concentrations of CO2, the absorption efficiency of the solution may be lowered to the extent that an appreciable increase in absorber height will probably be required. Peabody has developed a proprietary venturi-absorber that is insensitive to the H2S/CO2 ratio and is capable of handling fuel gas containing 85% CO2 with minimum absorption of CO2.8

The principal side reactions that are detrimental to the process are the formation of thiosulfates and thiocyanates. If hydrosulfate is contacted with oxygen before it is converted to sulfur, thiosulfate will form, the amount depending on the pH of the solution and on the operating temperature. The rate of thiosulfate formation under favorable conditions can be held to less than 1 wt % of the sulfur in the feed gas by controlling the pH and temperature of the solution and by assuring that sufficient delay time has occurred for the conversion of H<sub>2</sub>S to sulfur before the solution reaches the oxidizer tower. However, gases from direct-fired retorts contain oxygen and a large amount of  ${
m CO}_2$ . Calculations show that for those gases about 3 to 4% of sulfur in the gas will go to thiosulfate as the result of oxygen in the feed gas and the buffering action of the CO2. Fortunately, oil-shale gases contain only trace amounts of hydrogen cyanide (HCN). Hydrogen cyanide will react with elemental sulfur to form stable thiocyanide, which also accumulates in the solution.

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the wet sulfur cake is often disposed of by approved methods in a land-

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fill. Since the wet cake contains sufficient thiosulfate, additional purge of the solution is not required to control thiosulfate buildup. For plants producing more than 9 tonnes per day it may be more economical to recover the sulfur, in which case a small stream of solution must be continually purged.

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As thiosulfate is purged from the system it carries with it a proportionate amount of Stretford chemicals, which must be replaced (stream 18). It is important therefore to permit thiosulfate to concentrate in the solution to as high a level as the process will allow (20 to 22%) to minimize the costs of replacement chemicals and waste disposal.

The purge stream containing toxic vanadium salts with an average COD (chemical oxygen demand) of 20,000 mg/liter can be difficult to treat and dispose of. Treatment methods used include evaporation or spray drying, biological degradation, oxidative combustion, and reductive incineration. Peabody Holms has successfully developed a zero-discharge reductive incineration process that cracks the bleed liquor into a gas stream containing H<sub>2</sub>S and CO<sub>2</sub>, along with a liquid stream containing reduced vanadium salts that can be returned to the system. Set The problem of disposing of wastewater containing vanadium is eliminated and no makeup of vanadium or sodium salts is required. However, makeup water to the system must be demineralized and the feed gas to the Stretford absorber must be free of soluble minerals to prevent inorganic solids from building up in the system.

The Stretford process is insensitive to pressure and can operate in the range of atmospheric to maximum pipeline pressure. Operating temperatures range from ambient to  $49^{\circ}\text{C}$ . Any type of gas-liquid contacting device may be used as an absorber. Problems of plugging have been encountered in packed towers, especially with gases containing high concentrations of  $\text{H}_2\text{S}$  (above 1%). Therefore large rings or saddles are recommended when packed towers are used. Venturi scrubbers have been

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successfully used for gases containing larger amounts of H2S, up to 15%.<sup>10</sup>

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Stretford plants are remarkably free of corrosion tendencies and can be constructed entirely of carbon steel, with inert linings, e.g., coldcured epoxy resins, used for oxidizers and sulfur froth tanks. Stainless steel linings are recommended for solution and sulfur slurry pumps. 11

The Stretford process is extremely flexible in operation and can tolerate wide variations in both gas feed rate and H2S concentration in the feed gas. Startup and shutdown are relatively simple and can be accomplished in short periods of time. Good process control is maintained by simple analytical testing with little technical supervision. Since all process streams are handled as liquids, the process is easily automated and requires little operator attention.5

Hydrogen sulfide removal to less than 10 ppmv can be obtained with normal operation of a Stretford plant. Assuming that the gas contains COS in the range of 10 to 50 ppmv the overall sulfur removable achievable by the Stretford process would be in the range of 98.0 to 99.3%.

#### The EIC Process 2.

The EIC process was developed by EIC Corporation of Newton, Massachusetts, for removal of H2S from geothermal steam. The process has been field tested in a pilot plant processing 45,454 kg of steam per hour at The Geysers in California. 12 Although the process has not yet been pilot tested on desulfurization of fuel gases, EIC is promoting the process for that purpose.

Geothermal steam contains noncondensable gases such as carbon dioxide, hydrogen sulfide, ammonia, methane, nitrogen, hydrogen, traces of radon, argon, and mercury vapor, as well as rock dust, boron compounds, and other particulates. 12 In The Geysers test desulfurization to about 11 ppm  $H_2S$  in the exhaust steam was achieved. 12

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The EIC process is an absorption oxidation process using copper sulfate as the absorbent, which produces ammonium sulfate from hydrogen sulfide and ammonia in the treated gas. From the geothermal tests and from laboratory analysis EIC has found that the process has the following advantages: the process removes  $H_2S$  and COS but does not remove  $CO_2$ ; high removal efficiency of 99% is obtained in one absorption step; the systems can operate at elevated temperatures in the range of 66 to  $260^{
m oC}$ ; ammonia in the gas is removed and ends up in the by-product ammonium sulfate produced by the process; and the capital and operating

The chemistry of the EIC copper sulfate process is summarized as fol-

Scrubbing

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H<sub>2</sub>S Removal:

 $H_2S + CuSO_4 \longrightarrow CuS + H_2SO_4$ 

NH<sub>2</sub> Removal:

 $2NH_3 + H_2SO_4 \longrightarrow (NH_4)_2SO_4$ 

Regeneration

 $Cus + 20_2 \longrightarrow Cuso_4$ 

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Net Result

$$H_2S + 2NH_3 + 2O_2 \rightarrow (NH_4)_2SO_4$$

A flow diagram for the process is shown in Fig. 12. (stream 1) is scrubbed with a sulfuric acid solution of copper The raw gas at sulfate, which precipitates the hydroden sulfide as an insoluble copper sulfide and removes the ammonia by forming soluble ammonium sulfate. The precipitated copper sulfides (stream 2) are separated from the purge stream in a continuous decanter and are converted back to copper sulfate (stream 3) by leaching with oxygen. The purge stream (4) is neutralized with ammonia. The neutralized stream (5) is then treated intermittently with iron and hydrogen to remove any soluble copper for recycle to the process. The resulting effluent (stream 6), which is

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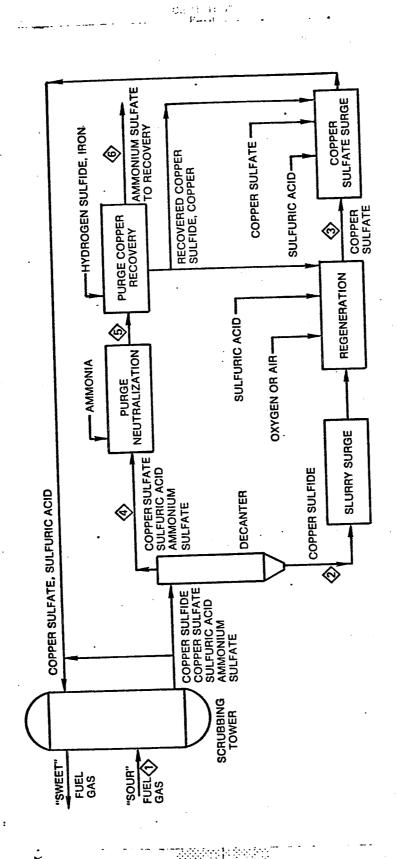


Figure 12. EIC Sulfuric Acid -- Copper Sulfate Process

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primarily ammonium sulfate, can be recovered by removing the water in a vacuum crystallizer. 13

Extensive use of titanium is required in the construction of the system because of the corrosiveness of the sulfuric acid—copper sulfate solution. 14 The size of the components, however, is much smaller than that of components for competing processes as the result of irreversible reactions occurring in the absorber and of the reactiveness and concentration of the solution. The smaller size of the equipment offsets a large part of the cost involved in the use of special materials of construction. EIC believes that the capital and operating costs of the system, including ammonium sulfate recovery equipment, are competitive with those of the Stretford process. 12

The process appears to be well suited to desulfurization of oil-shale gas since it removes both H2S and COS, as well as ammonia, and would not require that the feed gas be cooled. However, there are many unanswered questions, and more development work needs to be done before the process is applied to oil-shale gas. A potential major problem is that oil-shale gas contains various unsaturated hydrocarbons, which could react with the strong sulfuric acid-copper sulfate solution and form undesirable by-products. Of particular concern is the large amount of acetylene in the gas. If the highly unstable compound cuprous acetylide were to form and accumulate in the system, it could become very hazardous. Cuprous acetylide is explosive and can be detonated by percussion or can explode when heated above 100°C. If warmed in air or oxygen for several hours, it will explode when brought in contact with acetylene. 15 The potential for formation of cuprous acetylide in the EIC copper sulfate solution has not, to our knowledge, been defined.

Another potential problem could be entrainment of the solution in the treated gas. Carryover could cause corrosion of the downstream piping and gas turbine blades. The process has not been sufficiently proved \_\_\_\_\_\_\_to\_recommend its use at this time. The process could become a viable

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หลังได้เ ของหน candidate for desulfurization of oil-shale gases if through testing and development the adverse effects of corrosion and undesirable side reactions are proved not to create a problem.

### INDIRECT-RECOVERY PROCESSES

Table 35 gives the approximate volume and H<sub>2</sub>S concentration of the acid-gas stream that the model plant would produce for each of the indirect-recovery-process options. The selectivity data used for this table are either the average selectivity data reported in the literature <sup>16</sup>—<sup>18</sup> for the process or, as in the case for the MDEA<sup>19</sup> and the Diamox<sup>20</sup> processes, are the actual selectivity calculated for the process based on the gas composition assumed for the model plant. For the purpose of this comparison it was assumed that all H<sub>2</sub>S and a portion of the CO<sub>2</sub> relative to the average selectivity obtained by the process would be removed and end up in the acid-gas stream to the Claus unit.

Based on the requirements of a minimum acceptable volume of about 8% of H<sub>2</sub>S in the acid-gas feed to the Claus unit and of the need for at least 25% H<sub>2</sub>S for effective operation, the table indicates that a one-stage selective absorption process using DEA or MDEA would not produce a sufficient level of H<sub>2</sub>S in the acid-gas stream for operation of the Claus sulfur recovery process. Two stages of selective absorption using MDEA as the absorbent would be marginal, as would one stage of selective absorption using either the Diamox, the Benfield, or the Selexol process. Three stages of selective absorption using MDEA, however, would produce an acid gas rich enough in H<sub>2</sub>S for economic operation of the Claus unit.

It can be noted that the volume of acid gas produced decreases as the selectivity of the process increases. Thus when high selectivity is achieved, the volume of gas that must be processed by the succeeding absorption stage or by the Claus unit is considerably reduced, thereby reducing the equipment size and its installation and operating costs.

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Comparison of Selective Absorption Processes for Treating Gas from a Direct-Fired Retort Table 35.

Absorbent	Number of Stages		Acid-Gas Stream to Claus Unit					
		Selectivity	CO <sub>2</sub> (sm <sup>3</sup> d)	H <sub>2</sub> S (sm <sup>3</sup> d)	Total (sm <sup>3</sup> d)	Total (sm <sup>3</sup> d)	H <sub>2</sub> S (dry vol %)	
DEA	1	· 2 <sup>b</sup>	895,129	24,649	919,778	640	2.7	
MDEA	1	2.4 <sup>c</sup>	745,812	24,.649	770,467	535	3.2	
MDEA	2	8.2 <sup>c</sup>	219,466	24,649	244,172	170	10.1	
MDEA	3	27.4 <sup>C</sup>	65,316	24,649	89,994	62	27.4	
Diamox	1	4.7d	379,531	24,649	404,181	280	6.1	
Benfield	1	6 <sup>e</sup> .	299,782	24,649	323,016	224	7.6	
Selexol	1	9 <sup>£</sup>	198,921	24,649	280,170	-156	11.0	

Gas rate = 45,450 tonnes/day X 224 sm<sup>3</sup>/tonne = 10,188,000 sm<sup>3</sup>d; CO<sub>2</sub> = 17.572 vol % = 1,782,900 sm<sup>3</sup>d; H<sub>2</sub>S = 0.247 vol % = 24,649 sm<sup>3</sup>d.

b See ref 16.

C<sub>See ref 19.</sub>

d<sub>See ref 20.</sub>

e<sub>See</sub> ref 17:

f<sub>See</sub> ref 18.

The Benfield and Selexol processes, while only marginally capable of producing an acid gas of acceptable concentration, require high pressure to operate. The high cost of gas compression caused these processes to be categorically eliminated from further considerations (see discussion in Sect. VIII-B3).

The indirect-process candidates therefore were narrowed to a three-stage alkanolamine (MEDA) process and the Diamox process. A third candidate that was evaluated is a one-stage selective system coupled with a Stretford unit for recovery of sulfur from the acid gas. A discussion of these systems follows.

1. Three-Stage Selective Absorption Using MDEA as the Absorbent (See Appendix B).

Two principal commercial processes are available based on the use of amine solutions for selective absorption of  $\rm H_2S$  in the presence of  $\rm CO_2$ : the Selectamine process developed by Dow Chemical USA at Freeport, Texas, and the Adip process developed by Shell International Petroleum Company, The Hague, The Netherlands. The Adip process uses the secondary amine diisopropanolamine (DIPA) or the tertiary amine methyldiethanolamine (MDEA). The Selectamine process uses MDEA as the absorbent.

The basic flow steps for all alkanolamine acid-gas absorption systems are similar. As shown in Fig. 13, the gas to be purified (stream 3) enters the bottom of the absorber and passes upward countercurrent to the aqueous MDEA solution. The lean solution reacts with and chemically absorbs the H<sub>2</sub>S and a portion of the CO<sub>2</sub> as it contacts the gas in the absorber. The purified gas (stream 4) exits at the top of the absorber and the rich solution containing the absorbed acid gas (stream 6) is drawn off at the bottom. To achieve the highest practical selectivity, absorption kinetics are carefully controlled through absorber design and operation (see discussion in Sect. VII-B2a). The selectivities achieved and the approximate volumes of gas handled by each stage are given in Table 35.

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The sulfur removal effectiveness usually decreases as measures are taken to increase the selectivity. For the process described the  $H_2S$  content of the treated gas can be reduced to about 10 ppmv and the COS will be reduced by about 60%. The total sulfur reduction of the treated gas, assumed to contain 10 to 50 ppmv COS, would be in the range of 99.2 to 99.5%.

The rich solution (stream 6) is heated by heat exchange with the hot lean solution from the bottom (stream 5) of the stripping column and then enters the stripping column near the top. The lean solution partially cooled in the heat exchanger is further cooled by exchange with water and then fed into the top of the absorber.

In the stripper the absorption reactions are reversed as the temperature of the solution is increased. The desorbed acid gas exiting at the top of the stripping column is cooled to condense out the water vapor as the acid gas (stream 11) continues on to the next absorption stage. The condensed water is fed back at the top of the stripper above the point of the rich-solution entry. This serves to absorb the amine vapors carried out by the acid-gas stream.

The reaction between MDEA and organic sulfur compounds is not readily reversed. A side stream (7) of lean solution is sent to a reclaimer, where a portion of the degraded solution is recovered by high-temperature distillation. The bottoms (stream 10) accumulated from the reclaimer are sent to waste disposal.

The acid gas produced by the first-stage system ( $535 \text{ sm}^3/\text{m}$ ) becomes the feed for the next absorption stage. Compared to the gas entering the first-stage absorber (see Table -36) the quantity of gas that must be treated by the second-stage absorber has been reduced by 92% and the  $\text{CO}_2/\text{H}_2\text{S}$  ratio reduced to 30.8.

The second-stage absorber operates nearly identically to the firststage absorber. The treated gas (stream 12), however, is high-purity

Stage absorber. The treated ga

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Table 36. Performance of Three-Stage Selective Absorption
System Using MDEA

Stage	Feed Gas Volume (sm <sup>3</sup> m)	Acid-Gas Stream					
		Volume (sm <sup>3</sup> m)	Volume Reductiona (%)	H <sub>2</sub> S (%)	CO2 (%)	CO <sub>2</sub> /H <sub>2</sub> S	
						Ratio	% Reduction C
1	7,075	535	92.4	3.2	96.8	30.8	57.6
2	535	. 170	97.6	10.1	89.9	8.9	87.7
3	170	62.	99.1	27.4	72.6	2.6	96.4

Volume reduction related to volume of feed gas to first-stage absorber.

 $<sup>^{\</sup>rm b}{\rm CO}_2/{\rm H}_2{\rm S}$  ratio of feed gas to first-stage absorber is 72.6.

 $<sup>^{\</sup>mathbf{c}}\mathrm{CO}_{2}/\mathrm{H}_{2}\mathrm{S}$  ratio reduction related to feed gas entering first-stage absorber.

 ${\rm CO_2}$ , which contains about 10 ppmv  ${\rm H_2S}$  and is often used as a by-product when a local market exists for it.

The acid gas  $(170 \text{ sm}^3\text{m})$  (stream 17) produced by the second-stage system contains about 89.9%  $CO_2$  and 10.1%  $H_2$  and has a  $CO_2:H_2S$  ratio of about 8.9.

The third-stage absorber also produces high-purity  $\rm CO_2$  (stream 18). The acid gas (62 sm<sup>3</sup>m ) (stream 23), containing about 73%  $\rm CO_2$  and 27%  $\rm H_2S$  with a  $\rm CO_2:H_2S$  ratio of about 2.6, is rich enough in  $\rm H_2S$  content to be processed by a modified Claus system for sulfur recovery.

In the operation of the Claus unit (described in Sect. IV-B) the sulfur burning—by-pass process is used since it is the least costly option and since the acid gas produced by the three-stage selective absorption process is free of hydrocarbon contamination (see Fig. 14).

The SCOT tail-gas treatment was chosen because it can be integrated with the selective absorption system to save total installation costs. Since both systems use MDEA as the absorbent, the regeneration of the rich solution coming from the SCOT unit can be combined with regeneration of the absorbent from the primary absorber.

The SCOT process consists essentially of two parts: a reduction stage in which all sulfur compounds and elemental sulfur in the off-gas are reduced to  $\rm H_2S$  and an absorption stage in which, after water is removed by condensation,  $\rm H_2S$  is selectively removed by MDEA absorption/regeneration and is recycled to the Claus unit. The total sulfur recovery of the Claus unit can be increased to above 99.8% of the sulfur in the acid-gas feed. The SCOT unit is flexible, having a wide operating range, and no secondary waste streams are produced.  $^{21}$ 

The estimated overall sulfur reduction calculated for the total desulfurization train is about 98.8 to 99.3%, depending on the quantity of organic sulfur in the gas.

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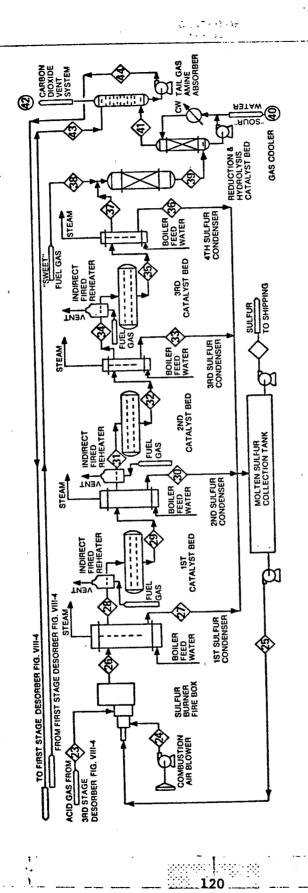
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Figure 14. Sulfur-Burning Claus Sulfur Recovery System with SCOT Tail-Gas Treatment Unit

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2. One-Stage Selective Absorption Followed by Treatment of Acid Gas by Stretford Process (see Appendix C)

As can be seen in Table 35 one stage of selective absorption us

As can be seen in Table 35 one stage of selective absorption using MDEA would produce an acid-gas stream containing approximately 3.2% H<sub>2</sub>S. This gas is too lean to be effectively processed with a Claus system. However, sulfur can be recovered from gases containing 3.2% H<sub>2</sub>S by the Stretford process. Thus a possible option would be to use one stage of selective absorption to remove the sulfur compounds from the gas followed by a Stretford unit to recover the sulfur from the acid gas.

Such a system (see Fig. 15 ) would have several advantages. The acid gas (stream 11) produced by the selective absorption step would be relatively clean, that is, free of hydrocarbons. Thus a much higher quality sulfur would be produced since many problems inherent with the Stretford process relative to impurities in the feed gas would be alleviated. The MDEA absorber would remove most of the organic sulfur compounds, COS, CS<sub>2</sub> and mercaptans, but they would pass through the Stretford absorber and end up in the CO<sub>2</sub> stream (16) discharged to the atmosphere.

The volume of acid gas (stream 11) fed to the Stretford unit would be reduced to about  $535~{\rm sm^3m}$  or about 7.6% of the original volume of the raw gas (stream 3). However, since stream 11 contains essentially the same amount of  $H_2S$  as stream 3, the amount of Stretford solution circulated; therefore except for the absorber the size of the Stretford equipment required would be the same as that for the Stretford direct-absorption case. Thus there is no cost advantage in this system over a Stretford direct system.

The estimated overall sulfur reduction calculated for the total desulfurization train is in the range of 98.0 to 99.3%, depending on the quantity of organic sulfur in the gas.

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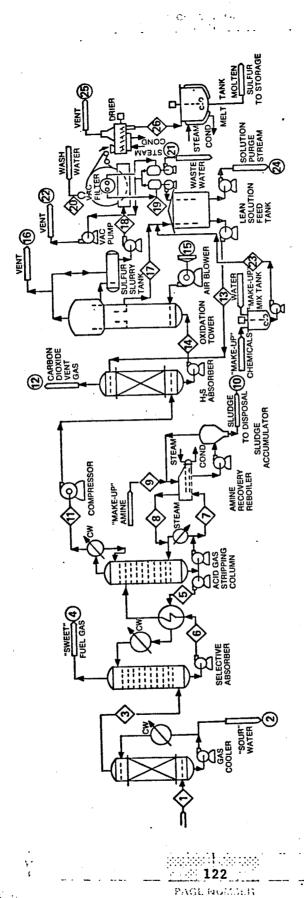
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One-Stage MDEA Selective Absorption System with Stretford Sulfur Recovery Unit Figure 15.

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3. The Diamox Process (See Appendix D)

The Diamox process is a commercially proven process developed jointly by Mitsubishi Chemical Industries (MCI) and Mitsubishi Kakoko Kaisha (MKK), both of Japan, for desulfurizing coke-oven gas. Five facilities employing the process have been built, and their performance has been excellent. MCI reports<sup>20</sup> that the Diamox plants have achieved coke-oven gas desulfurization to levels as low as 13.7 to 18.3 grams of H<sub>2</sub>S per 100 sm<sup>3</sup> (about 95 to 126 ppmv). The Ralph M. Parsons Company of Pasadena, California, is the licensee for the process in the United States.

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The Diamox process is particularly applicable for removal of H<sub>2</sub>S from gases containing ammonia since ammonia is used as the absorbent in the Diamox solution. Ammonia seems in the raw gas is absorbed to generate the Diamox solution, and consequently no chemicals are required for the process. Once the Diamox solution is saturated with ammonia, additional ammonia will not be absorbed but instead will pass through with the treated gas. Additional ammonia scrubbers must be added if ammonia has to be removed from the treated gas.

A large part of the proprietary design of the Diamox process is embodied in the design and operation of the absorber and stripping towers and the temperature and solution concentrations maintained throughout the system. 22 The process was developed to selectively absorb H<sub>2</sub>S from coke-oven gas with greatly improved removal effectiveness. Up to 98% H<sub>2</sub>S removal has been obtained in commercial applications of the process compared to removals of only 90% achieved by conventional processes using ammonical solutions. Selectivity ratios of up to 9.4 have been obtained by the process based on data reported by Hiraoka, Tanaka, and Sudo. 23 Organic sulfur compounds, however, are not appreciably removed.

High-quality, 99.9% pure, bright-yellow sulfur can be recovered from the acid gas produced by the Diamox process since the concentration of

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ammonia, hydrocarbons, and other contaminants in the acid gas is extremely low.

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A simplified flow diagram of the process is shown in Fig. 16. Diamox process operates as follows. The incoming raw gas (stream 1) is cooled to about 52°C by direct contact with circulating water in a wash tower. Water, with a small amount of absorbed ammonia, and a small quantity of hydrocarbons condensed from the gas are purged from the system (stream 2) and sent to foul-water treatment. Cooling to 52°C permits the bulk of the ammonia to pass through with the cooled The cooled gas (stream 3) then enters the H<sub>2</sub>S absorber, where it countercurrently contacts the freshly stripped, lean ammoniacal solution. Nearly all the H<sub>2</sub>S and a portion of the CO<sub>2</sub> are absorbed before the treated gas (stream 4) exits from the top of the absorber. rich solution (stream 5) enters the acid-gas stripper, where the absorbent solution is heated to expel the absorbed acid gases. stripped or lean absorbent (stream 6) is cooled and then returned to the H<sub>2</sub>S absorber; the stripped acid gas (stream 9) is sent to the Claus unit for recovery of sulfur.

A small amount of solution (stream 7) is purged from the recirculating absorbent solution to control buildup of impurities and then is sent to foul-water treatment.

The goal of obtaining a high degree of sulfur removal conflicts with that of simultaneously obtaining high  $H_2S$  selectivity; one goal must be compromised to obtain the other. To obtain a high degree of desulfurization the partial pressure of  $H_2S$  in the absorbent must be low and the acid-gas concentration in the regenerated lean solution must be at the lowest possible level. Thus large amounts of solution must be circulated through the absorber and greater amounts of utilities consumed in regenerating and recirculating the solution.

The system proposed for the model oil-shale gas disulfurization plant \_\_\_\_\_ OMERSION is\_based on a spray tower absorber with six absorption stages (see \_\_\_\_\_: FORTABLE)

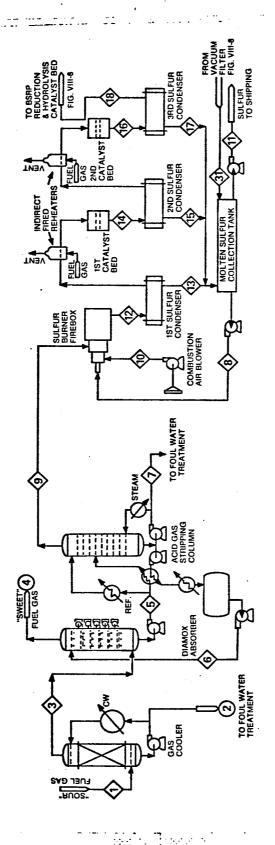
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Appendix D).<sup>20</sup> The high CO<sub>2</sub> content results in increased CO<sub>2</sub> absorption. Calculations indicate that the H<sub>2</sub>S content of the treated gas will be about 63 ppmv and that the acid gas stripped from the rich Diamox solution in the acid-gas stripper contains about 6 mole % H<sub>2</sub>S.<sup>20</sup> This equates to an H<sub>2</sub>S removal of 97.9% while achieving a selectivity ratio of 4.7. With the gas assumed to contain COS in the range of 10 to 50 ppmv the overall sulfur removal achievable would be 96.3 to 97.6%. The acid gas is fed to a Claus sulfur recovery unit for conversion of the H<sub>2</sub>S to elemental sulfur (Fig. 16 ). The Claus unit tail-gas feeds to a Beavon sulfur removal process (BSPR) for final sulfur recovery (Fig. 17 ). The BSPR unit was chosen due to the large amount of CO<sub>2</sub> in the acid gas.

The Beavon sulfur removal process for treatment of Claus plant tail gases consists of two main steps: catalytic hydrogenation and hydrolysis of all sulfur species to  $\rm H_2S$  and conversion of the  $\rm H_2S$  to elemental sulfur by the Stretford process.

The overall sulfur recovery in the Claus and BSRP units is 99.8% percent.<sup>20</sup> The estimated overall sulfur reduction calculated for the total desulfurization train is about 96.1 to 97.4%, depending on the quantity of organic sulfur in the gas.

The Diamox process, while suitable for coke-oven gas containing large amounts of ammonia, carbon dioxide, and hydrogen cynide, is only marginally capable of processing oil-shale gas because the ratio of  ${\rm CO}_2$  to  ${\rm H}_2{\rm S}$  in the gas is very high. The process would not be capable of producing an acid gas of sufficient  ${\rm H}_2{\rm S}$  content for effective operation of the Claus system for most direct-fired oil-shale retort gases without considerable sacrifice of  ${\rm H}_2{\rm S}$  removal effectiveness.

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BSRP Tail-Gas Treatment Unit Figure 17.

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\*When a reference number is used at the end of a paragraph or on a heading, it usually refers to the entire paragraph or material under the heading. When, however, an additional reference is required for only a certain portion of the paragraph or captioned material, the earlier reference number may not apply to that particular portion.

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# X. COST COMPARISON OF CANDIDATE PROCESSES

Estimated capital and operating costs and cost-effectiveness data for the candidate processes described in Sect. VIII are presented in this section. The estimates were based on a hypothetical direct-fired oilshale plant processing 45,450 tonnes of shale per day and producing 10.2 million sm<sup>3</sup>d of gas having the composition shown in Table 34 Material balance flow sheets were developed for each of the selected candidate processes. The major equipment components were defined and sized and the capital and operating costs estimated.

The estimated capital and operating costs and net annualized cost of the various options evaluated are given in Table 37. The material balance flow sheets and a summary of the equipment requirements and cost estimate details for each process are given in Appendices A, B, C, and D.

The estimated costs are based on a new-plant installation and represent the total investment, including all indirect costs such as engineering and contractors' fees and overheads, required for the purchase and installation of all equipment and material to provide a facility as described. These are battery-limit costs and do not include provisions for bringing utilities, services, or roads to the site; backup facilities; land; required research and development; or process piping interconnections that may be required within the process that generates the gas fed to the desulfurization systems.

Capital cost estimates were developed by summation of installed capital costs for the individual components of each system. These installed capital costs are based on IT Enviroscience experience adjusted to the January 1980 basis. In addition to the sum of the itemized capital costs a contingency allowance of 20% is included in overall capital cost estimates.

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Table 37. Estimated Capital and Operating Costs for Fuel-Gas Desulfurization Options a

			Costs	i	·
<b>I</b> tem	Stretford Stretford Direct Direct Process with Process with Purge Stream Disposalb Recovery		1-Stage MDEA Selective Absorption with Stretford Sulfur Recovery	3-Stage MDEA. Selective Absorption with Claus Sulfur Recoverye	Diamox Process With Claus Sulfur Recovery
Capital cost <sup>g</sup>	\$13,322,000	\$14,840,000	\$17,200,00	\$15,551,000	\$32,741,000
Operating cost (per year)				<b>i</b> :	
Chemical makeup	\$1,088,000	\$ 911,000	\$ 1,190,000	\$ 24,000	\$ 84,000
Steam	70,000	107,000	3,812,000	7,183,000	11,530,000
Power .	303,000	345,000	234,000	158,000	1,618,000
Fuel gas		328,000		;	
. Cooling water	526,000	578,000	1,261,000	1,545,000	1,950,000
Waste treatment	No cost	Neg.	No cost	Neg.	No cost
Operating labor	600,000	720,000	630,000	480,000	480,000
Maintenance, capital recovery and misc.	3,864,000	4,304,000	4,988,000	4,510,000	9,495,000
Total operating cost	\$6,451,000	\$7,293,000	\$12,115,000	\$13,900,000	\$25,157,000 :
Sulfur recovery credit (per year)	\$1,083,000	\$1,116,000	\$1,130,000	\$1,171,000	\$1,150,000
Net annualized cost		•			
Per year	\$5,368,000	\$6,177,000	\$10,985,000	\$12,729,000	\$24,007,000
Per ton of shale	0.29	0.34	0.60	0.70	1.32
Per barrel of oil	· 0.50	. 0.57	1.01	1.17	2.21

For 360-million-scfd plant based on Paraho gas (see Table VIII-1). Stretford purge stream disposal at no cost.

Stretford purge stream incinerated and gases, water, and decomposed sodium salts recycled. Sulfur recovered with Stretford system. Sulfur recovered with Claus system with SCOT tail-gas treatment. Sulfur recovered with Claus system with BSRP tail-gas treatment.

The actual cost of applying any of these options at any specific location could vary considerably from that given here. The purpose of these estimates was to determine the relative cost differences between the various options. Since the estimates for each of the options evaluated were calculated on the same bases, the relative costs of applying the systems should be in the order shown.

17. A. B.

Special cautions must be used if an attempt is made to extrapolate from the cost-effectiveness data given since these costs will be largely dependent on the composition and volume of gases to be treated, on the oil yields, and on the cost factors that are pertinent to installation site.

For desulfurization of gases from direct-fired oil-shale retorts the Stretford process is the most cost-effective system. For the model case shown the cost of sulfur removal would be about \$0.50 per barrel of oil produced, which is less than half that projected for the better of the indirect sulfur-removal processes. The model case is based on Paraho gas, which has the lowest CO2/H2S ratio (76:1) of any gas considered for the direct-fired retorts. For direct-fired gases having higher CO<sub>2</sub>/H<sub>2</sub>S ratios the gap between the cost of applying the Stretford process and the indirect desulfurization processes would be expected to increase, making the other options even less competitive. For gases with lower CO2/H2S ratios the indirect sulfur-removal processes would become more competitive. For gases from indirect-heated retorts, such as Tosco with a CO2/H2S ratio of 4.1:1, the indirect processes could very well become more cost effective than the Stretford process.

The operating costs given are based on the assumption that there is no cost for wastewater treatment. The sour water stream in each case would be stripped of H2S and processed for recovery of hydrocarbon and ammonia by-products. The cost of this treatment, which is partly offset by the valve of the by-products recovered, was assumed to be equal in each case and therefore is not included. ... The Stretford purge

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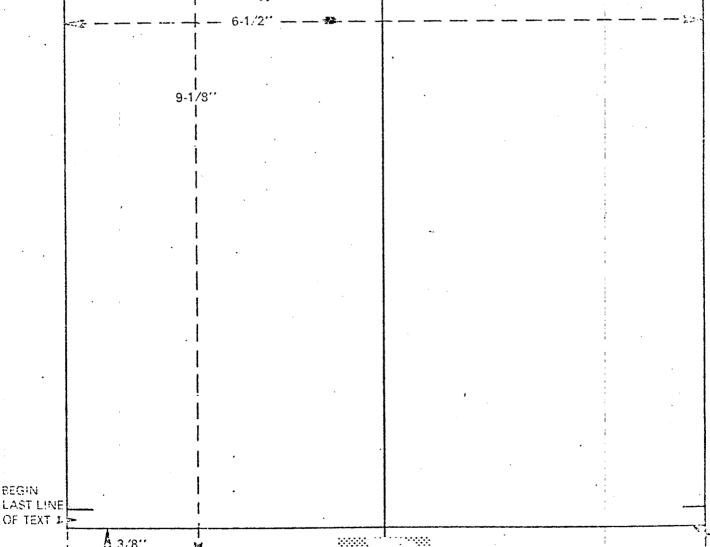
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streams, containing essentially dissolved salts, were assumed to be acceptable for disposal through moisturizing of the spent shale at no The cost of evaporating the purge streams to recover dissolved salts would be about \$557,000/yr (\$0.05/bbl) for the Stretford direct systems and about \$37,000/yr (\$0.003/bbl) for the BSRP system used for treatment of the tail gases from the Diamox process Claus unit.

The cost of a Stretford system incorporating a purge-stream reductiveincineration system for recovery and recycle of salts is also given in Table IX-1. This option would result in a sulfur removal cost of about \$0.57 per barrel of oil produced, which is about \$0.07 more per barrel than the case based on disposal of the purge stream at no cost.



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The Stretford direct fuel-gas desulfurization process was recommended as the most applicable technology for removal of sulfur from direct-fired oil-shale retort gases. The basis for this selection is as follows.

#### A. ADAPTABILITY

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Gases from direct-fired retorts have a low heating value (1.52 to 4.13x10<sup>6</sup> joules/sm<sup>3</sup>) because of the large amount of nitrogen, carbon dioxide and other inert gases in the gas. It is not economically feasible to upgrade the quality of the gas so that it will meet pipe-line quality standards and can therefore be sold as a substitute for natural gas. Practical uses of the gas are limited to combustion for generating process heat or electrical power, for which purpose only the ammonia and sulfur need to be removed.

The large volumes of gas that must be processed in a typical oil-shale plant will limit the application of desulfurization technology to high-capacity, continuous-liquid-phase processes. Since CO<sub>2</sub> is absorbed to some extent by all liquid-phase processes, the high CO<sub>2</sub>/H<sub>2</sub>S ratio of the gas limits the selection to those processes that can selectively absorb sulfur compounds in the presence of large amounts of CO<sub>2</sub>. Of those direct processes that selectively remove H<sub>2</sub>S by direct conversion of the H<sub>2</sub>S to elemental sulfur, the Stretford process is the most effective. Of those indirect processes that selectively remove H<sub>2</sub>S by separating the H<sub>2</sub>S as a concentrated acid-gas stream, the Selectamine or the Adip processes using MDEA as the absorbent, the Benfield, the Selexol, and the Diamox processes were selected as the most effective in their separate process classifications. These processes were than considered as candidates for further study.

### B. OPERATIONAL REQUIREMENTS

Oil-shale retort gases are produced at near-atmospheric pressures; thus those processes requiring the gas to be at a high pressure (Benfield

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The Claus process is used to recover sulfur from the acid gas produced by the indirect sulfur-removal processes. The large amount of CO<sub>2</sub> in the gas makes the best of the indirect process only marginally capable of producing an acid gas rich enough in H<sub>2</sub>S for processing by the Claus process. Thus to apply these processes, multiple stages of selective absorption would be required to handle the gas produced by many of the direct-fired retorts.

The Stretford direct process, on the other hand, is only minimally affected by the quantity of CO<sub>2</sub> in the gas and therefore is adaptable to the full range of gases produced by direct-fired retorts.

A Stretford plant incorporating a venturi absorber can operate with a wide-volume turndown range. The system capacity is limited by the maximum design-basis throughput capacity of the sulfur. The quantity of sulfur in the gas can drop to nearly zero without affecting the sulfur removal efficiency.

The indirect selective sulfur-removal processes have only limited turn-down capability. The selectivity of the process (preferencial absorption of  $\rm H_2S$  over  $\rm CO_2$ ) is largely achieved through control of the absorption kinetics. Therefore large changes in absorber gas or liquid throughput can adversely affect the selectivity achieved and consequently the composition of the acid gas produced. The Claus process has a limited range of operation and can be upset by large variations in the composition of the acid-gas feed. The wide range and constant variation of the  $\rm CO_2/H_2O$  ratio of gases from in-situ retorts would be very difficult to handle with an indirect selective absorption system.

#### C. SULFUR-REMOVAL EFFECTIVENESS

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A very high degree of H<sub>2</sub>S removal can be achieved by all the candidate processes discussed. Generally the higher the amount of sulfur that...

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must be removed the more it will cost to install and operate the desulfurization train. The cost of applying these systems then becomes the overriding factor rather than the ultimate sulfur-removal effectiveness.

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Except for the Diamox process all the candidate processes are capable of removing H<sub>2</sub>S down to about 10 ppmv. However, organic sulfur compounds, principally COS, which exists in only trace amounts in the gas, are not significantly removed or are only partly removed by the various process. Table 38 gives the variation in process effectiveness with the amount of COS in the raw gas. Current Colorado emission standards limit sulfur emissions to an equivalent of 0.3 lb of SO<sub>2</sub>/bbl of crude oil produced. The estimated SO<sub>2</sub> equivalent emissions for each of the candidate process are shown in Table 38.

#### D. RELATIVE COSTS

The relative costs of the candidate processes studied are given in Table 39.

For desulfurization of gases from direct-fired oil-shale retorts the Stretford process is the most cost-effective system. The cost ratios shown, based on Paraho gas, would be expected to increase for other direct-fired-retort gases having higher  $\rm CO_2/H_2S$  ratios.

The three-stage MDEA process produces the highest degree of sulfur recovery, whereas the Stretford process produces the lowest. In the latter process sulfur recovery losses result from formation of sodium thiosulfate in the Stretford solution.

#### E. WASTES GENERATED

The Stretford process produces a concentrated waste stream containing about 22% dissolved solids. The MDEA selective absorption systems produce a small quantity of concentrated sludge from the solution reclaimer. The quantities of these wastes are as follows:

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Table 38. Variation of Process Effectiveness with Amount of COS in Raw Gas<sup>a</sup>

			Total Sulfur	Sulfur	:	Estimated Overall Emissions	rall Emissions		Sulfur
	n2a Beninian	Bedistrion	Reduction	Remaining	Tota	Total Sulfur	SO2 Equ	SO <sub>2</sub> Equivalent	Reduction
	Remaining (nomy)	(vt %)	(vt %)	(kg/hr)	(kg/hr)	(tonnes/day)	(kg/hr) (1	(kg/hr) (kg/bbl of oil)	(vt %)
Stretford	1	Ni1	98.0 - 99.3	19 - 58	19 - 58	0.45 - 1.35	38 = 115	0.02 - 0.05	£'66 - 0'86.
MDEA - 1 stage	10	09	99.2 - 99.5	13 - 29	20 - 59	0.45 - 0.90 40 - 117	40 - 117	0.02 - 0.05	98.0 - 99.3
with Stretford MDEA - 3 stages	10	9	99.2 - 99.5	13 - 29	20.4 - 36	0.45 - 0.90	41 - 72	0.02 - 0.04	98.0 - 99.3
with Claus/SCOT	£9	Nil	96.3 = 97.6	70 - 109	75 - 114	1.81 - 2.70 151 - 228	151 - 228	0.07 - 0.11	96:1 - 97.4
Claus/BSPR									

abased on a Paraho plant producing 8,000 m3 of oil per day with COS ranging from 10 to 50 ppmv in the raw gas.

Table 39. Relative Costs of Various Gas Desulfurization Options a

			1-Stage MDEA	3-Stage MDEA	
	Stretford	Stretford	Selective	Selective	Diamox
•	Direct Process	Direct Process	Adsorption	Absorption	Process
	with	with	with	with	with Claus
	Purge Stream	Purge Stream	Stretford Sulfur	Claus Sulfur	Sulfur
Category	Ulsposal	Recycle	Recovery	Recovery	Recovery
Installed capital cost	1.00	1.11	1.29	1.17	2.46
Utility costs	1.00	1.51	5.90	9.88	16.79
Total operating costs	1.00	1.13	1.88	2.15	3.90
Value of sulfur recovered	0.92	0.95	96*0	1.00	0.98
Net annual cost	1.00	1.15	2.05	2.37	4.47
Cost effectiveness	1.00	1.15	2.04	2.34	4.54

Lost effectiveness

Lost effectiveness

Based on a plant processing 10.2 million sm<sup>3</sup> of Paraho gas per day. The most effective or least costly option in each category is shown as unity (1.00). The relative costs of the other options are shown as a ratio to the most effective

Stretford direct
3-stage MDEA with Claus/SCOT
1-stage MDEA with Stretford
Diamox with Claus/BSRP

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7570 L/day 37.85 L/day 7570 L/day 246 L/day

These estimates do not include sour water, which would be processed for hydrocarbon and ammonia recovery prior to disposal. The quantity of sour water in each case would be about the same  $(1.13 \times 10^6 \text{ to } 1.51 \times 10^6 \text{ l/day})$ .

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#### F. RELIABILITY

All the candidate processes were commercially developed and are used in various applications around the world. The stretford process has been widely applied, with more than 80 commercial plants currently in operation.

As with any process application operating problems can develop, and if the system is misapplied or poorly designed, the problems can be insurmountable. The principal problems reported by Stretford users are plugging of packed absorbers by precipitated sulfur; loss of scrubbing performance, when processing gases with large percentages of CO<sub>2</sub>, because CO<sub>2</sub> is absorbed and the solution pH is consequently lowered; and poor-quality sulfur as the result of hydrocarbon and other contaminants in the sulfur.

The process licensors feel confident that with proper design and through use of proprietary Stretford chemicals these problems can be managed.

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#### XI. PILOT-PLANT DESIGN

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#### INTRODUCTION A.

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Gases produced from direct-fired oil-shale retorting processes are sufficiently different from gases previously encountered in commercial application of gas-desulfurization systems that the technology cannot be simply transferred. Application of the Stretford process or any other process to the treatment of these gases would extend the technology of the process into areas in which no analogous experience is available. Many questions need to be answered before the process can be applied with confidence to a full-scale commercial shale-oil production facility.

The primary function of the pilot plant would be to test the technical feasibility of the Stretford process for treatment of gases from oilshale retorting processes, to prove the engineering assumptions and calculated forecasts made for the process, and to generate scaleup design data for commercial applications of the process.

The pilot plant should be portable and capable of operating on a slip stream from any of the currently operating or proposed direct-fired oil-shale retorting facilities in the United States. The unit should be flexible enough to accommodate inputs from surface and in-situ retorts with flow rates ranging from 2.83 to 28.3 sm<sup>3</sup>m while retaining adequate design efficiencies; if possible it should have sufficient turndown capability to also operate on gases from indirect-heated retorts.

The pilot plant would be a Stretford system based on the current stateof the-art technology for commercial applications of the process. unit should be the smallest size that will operate on actual shale-oil retort-gas feeds under typical plant constraints and still permit study of the process dynamics.

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#### B. SIZING OF PILOT PLANT

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MEAD: PEGALL Based on the gas characterization data compiled the  $\rm CO_2/H_2S$  mole ratios are widely separated in the gases from indirect- and direct-fired retorting processes, and it is unlikely that a fixed-size pilot plant could handle the full range of gases from both types of retorts. With direct-fired retorts, either above or below ground where combustion occurs within the retort, gas rates vary from 68.5 to 405 sm<sup>3</sup>/tonne of shale processed and the  $\rm H_2S$  content varies from 0.07 to 0.30 vol % (DG). The  $\rm CO_2/H_2S$  molar ratio varies from 76 to 165 or more.

For indirect-heated retorts, where heat to the retort is generated externally, the gas rates vary from 27.2to  $35.0~\rm sm^3/tonne$  of shale processed and the H<sub>2</sub>S content varies from 3.8 to 4.1 vol % (DG). The CO<sub>2</sub>/H<sub>2</sub>S molar ratio varies from 4.3 to 5.0.

Thus for a volume turndown ratio of 10 (28.3 to 2.83 sm $^3$ m) the pilot plant would have to have a sulfur turndown capability of 586 to 1 if the H $_2$ S content of the gas ranged from 0.07 to 4.1 vol %. If the plant were designed for only gases from direct-fired retorts, a sulfur turndown capability of 34 to 1 would be required. Similarly if the plant were designed for only gases from indirect-heated retorts, a sulfur turndown capability of 11 to 1 would be required. However, if the gas volume were decreased as the sulfur content was increased, then a sulfur turndown capability of only 6 would be required to handle the full range of gases from both direct and indirect oil-shale retorts (i.e., 2.83 sm $^3$ m of gas from an indirect-heated retort containing 4.1% H $_2$ S to 28.3 sm $^3$ m of gas from a direct-fired retort containing 0.07% H $_2$ S).

In reality the Stretford system is limited in both gas throughput capacity and sulfur loading capacity. Therefore to provide maximum turndown capability, the pilot plant should be sized to handle the maximum case for direct-fired retorts  $(28.3~{\rm sm}^3{\rm m})$  of Paraho gas containing 0.30 vol %  ${\rm H}_2{\rm S}$ ). Gases containing less sulfur can be handled up to  $28.3~{\rm sm}^3{\rm m}$  since the system is capable of operating with reduced sulfur

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loadings. For gases containing more sulfur, the gas volume can be turned down to within the limits of the absorber operation while maintaining maximum sulfur loading.

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The pilot plant should be designed primarily to remove  $\rm H_2S$  from oilshale gas produced by direct-fired retorts. The maximum capacity of the pilot plant should be 28.3 sm<sup>3</sup>m of feed gas. The maximum sulfur capacity should be 6.6 Kg of sulfur per hour based on a loading of 800 ppmw of hydrosulfide ion (HS<sup>-</sup>) in the solution. The plant should be capable of reducing the  $\rm H_2S$  content of the gas to 10 ppmv or less, and  $\rm CO_2/H_2S$  ratios as high as 200 to 1.

The pilot plant should include equipment for cooling the feed gas from 60° to 32°C and for removing the ammonia before the feed gas enters the Stretford absorber. The pilot-plant design should include only those equipment components, instruments, and controls necessary for safe operation of the equipment and as required to adequately test and demonstrate the Stretford process and to carry out the intended research program. Most of the data required for determining the performance and operation of the system would be obtained by manual sampling and laboratory analysis of the process streams.

- D. RECOMMENDED SYSTEM
- 1. Basis of Design

The bases for designing and sizing the pilot-plant components are as follows:

Gas characteristics

Maximum volume (dry gas)

28.3 sm<sup>3</sup>m

Pressure

Atm

Temperature

60°C

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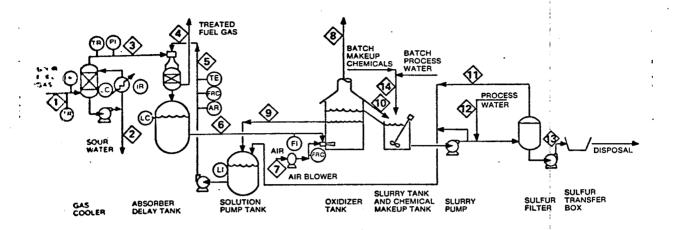
Critical components 0.3 vol % H<sub>2</sub>S 22.8 vol %  $CO_2$ 0.7 vol %  $NH_3$ ·· Saturated 21°C Cooling water temperature Not included Sour-water treatment Disposal as wet cake Sulfur recovery Stretford solution sulfide loading 800 ppmw Description of Process 2. A material balance flow sheet for the Stretford pilot plant is shown on Fig. XI-1. The idealized process reactions are as follows: Reaction 1  $9.1 \text{ Na}_2\text{CO}_3 + \text{H}_2\text{S} \longrightarrow \text{NaHS} + \text{NaHCO}_3$ Reaction 2  $\rightarrow$  Na<sub>2</sub>V<sub>4</sub>O<sub>9</sub> + 4NaOH + 2S 4NaVO3 + 2HaHS + H2O -Reaction 3  $Na_2V_4O_9 + 2NaOH + 2ADA^* \longrightarrow 4NaVO_3 + 2ADA (reduced)$ Reaction 4 2ADA (reduced) +  $O_2 \longrightarrow 2ADA^* + H_2O$ \*Anthraquinone disulfonic acid. The sour fuel-gas feed (stream 1) enters the gas cooler, where it is cooled by direct contact with circulating water solution. The flow BOTTOM OF rate of feed gas is controlled to the rate set for the test and is continually recorded. As the gas is cooled, a large portion of the LAST LILE OF TEXT 5 water\_and most\_of the ammonia are removed (stream 2)....Heat is ex-

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ره <u>.                                    </u>	110.68		11068	110.68	Water	20000.00	20006.00	2.80	4.80	19832.00	168.00	92.13	8.79		5.81
CD.	1586.67	48.67	1538.00	1538.00	Sulfur		8				14.60	<u> </u>		24.60	
CH.	54 72		54.72	54.72	Vanadate	114.90	114.90			113.90	0.96	0.65		0.03	0.0
ζ, ñ,	29 28		29.28	29.28	ADA	214.90	214.90			213.10	1.80	1.02		0.06	0.0
C. H.	26.61		26.61	26.61	EDTA	54.00	54.00			53.50	0.45	0.25		0.02	0.0
Z. 44.	48.38		48.38	48.38	Iron (Soluble)	2.00	2.00			1.98	0.02	0,01	1	Trace	Trac
7 H					NaHCO,	1016.00	1055.60			1046.70	8.86	5.53		0.31	
~ 4 4 <del> </del>	50.55		50.55	50.55	Na, CO,	59.90	10.00			9.92	80.0	0.23		Trace	0.2
C 1	68 54	Trace	68.54	68.54	Na, S, O,	3822.90	3824.00			3791.90	32.10	18.42		1.11	<u> </u>
4,1	_	of H-C's													<u> </u>
M. S	16.17	0.12	16.05	00.54	Hydrocarbons		Trace		Trace						<u> </u>
400	18 83	18.83	Trace	_								1			
<b>₩</b> ,.5	712.03	581.87	130.16	130.16											
13,															1
Fi Car															
Y	55€5 38	649.49	4915.89	4899.89	Total, lb/hr	25285.00	25301.00	160.40	155.10	25063.00	226.87	118.25	8.79	40.73	6.2
THE WINDS	1000 00		986.00		Total, SCFM			34.50							
Yang	140' F	90°F	90°F		Total, gpm	42.20									
	1					<del> </del>									

Material Balance Flowsheet for Pilot Plant

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tracted by circulating the water solution through a water-cooled heat exchanger.

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The temperatures of the incoming sour gas, the cooled gas, and the cooling water discharge streams are continually monitored. The sour water (stream 2) is discharged. Treatment of the sour water is outside the scope of the pilot plant.

The cooled gas (stream 3) then enters the ejector-venturi gas absorber at about 90°F. The ejector-venturi absorber (see Appendix E) operates on the jet principle, in which the Stretford solution is under pressure to create a draft and simultaneously provide intimate contact between the liquid and gas as it passes through the venturi. On contact with the alkaline Stretford solution, which is an aqueous solution containing sodium carbonate, sodium metavanadate, and anthraquinone difulfonic acid (ADA), the H<sub>2</sub>S dissolves rapidly and forms a small concentration of hydrosulfide ion (reaction 1). Sodium carbonate in the solution enters into the reaction and provides a buffer to prevent rapid pH changes as the acid gases are absorbed.

The solution will absorb some  $\mathrm{CO}_2$ , resulting in formation of sodium bicarbonate and consequent lowering of the pH until equilibrium is reached with respect to the level of  $\mathrm{CO}_2$  in the gas. The pH of the Stretford solution will therefore be a function of the level of  $\mathrm{CO}_2$  in the gas. At the point when equilibrium is reached, there must be sufficient  $\mathrm{Na}_2\mathrm{CO}_3$  remaining to convert all the  $\mathrm{H}_2\mathrm{S}$  in the solution to hydrosulfide ions.

For gases with large  $\mathrm{CO}_2/\mathrm{H}_2\mathrm{S}$  ratios the pH of the solution may be sufficiently lowered to affect the absorption rate of  $\mathrm{H}_2\mathrm{S}$ . In that case addition of proprietary additives to the solution may be required for the hydrosulfide loadings to be as high as 800 ppm.

The venturi discharges into a mist eliminator, where gases are separated from the liquids. The treated gas (stream 4), in which essentiated

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tially all the  $H_2S$  has been removed, will be returned to the oil-shale process. The separated liquid will flow into the delay tank.

The sodium metavanadate in the solution readily reacts with the hydrosulfide to produce elemental sulfur (reaction 2) and a reduced form of vanadium. Sufficient residence time is provided by the delay tank to assure that any remaining hydrosulfide in the solution is reacted so as to prevent hydrosulfide ions from entering the oxidizer and becoming oxidized to sodium thiosulfate, an undersirable, stable, by product.

Flow (stream 6) from the delay tank is by gravity to minimize sticking of the sulfur when the solution is in a transitory Redox stage. A liquid-level controller is provided to control the flow of solution from the delay tank to the oxidizer. The reacted solution containing reduced vanadium and reduced ADA is regenerated in the oxidizer tower according to reactions 3 and 4 by being sparged with an excess of air (stream 7) supplied by a blower. The flow of air can be adjusted so that sulfur particles will properly rise, forming a froth at the surface but allowing the Stretford solution to settle clear of sulfur. The accumulated sulfur froth overflows (stream 10) to a slurry tank. Trace amounts of volatile hydrocarbons that may be absorbed in the Stretford solution will be stripped by the air rising in the oxidizer. A hood provided over the oxidizer tank will lift and direct the discharge gases (stream 8) away from the operators.

The regenerated Stretford solution (stream 9) is essentially free of sulfur and is returned to the solution pump tank. The solution pump tank is sized to hold all the required solution when the pilot plant is shut down.

The regenerated solution (stream 4) is then pumped back to the venturi absorber at a preset and controlled rate. The flow rate, pH, and temperature of the solution are continually monitored and recorded.

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Equipment for handling the sulfur slurry has been reduced to the minimum in an attempt to keep the cost of the pilot plant to a minimum and yet provide sufficient capability to test and demonstrate the Stretford process.

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The sulfur produced will be disposed of as a wet cake. A sulfur filter (see Appendix E) operated on a batch basis will filter the sulfur from the slurry. The filtrate solution (stream 11) is returned to the solution pump tank. The wet sulfur cake accumulated in the filter will be discharged to a transfer box, which is sized for 4 days of operation at maximum utilization. The amount of Stretford solution lost with discharge of the wet sulfur cake (stream 13) should be sufficient to keep the buildup of sodium thiosulfate in balance. The cost of Stretford chemicals lost in the sulfur cake at the pilot level of operation is not a significant economic factor. The filter system is arranged so that the filter cake can be rinsed (stream 12) before is is discharged.

#### 3. Turndown Capability

The described capacity of the pilot plant (6.6 Kg of sulfur/hr) is based on an average solution loading of 800 ppm. Sulfur loadings of 500 to 1000 ppm are commonly achieved. The maximum solution loading obtained will depend on many factors but will largely be a function of the solution pH, which is influenced by the amount of CO2 in the gas. Solution loadings can be anywhere from zero up to the maximum capacity of the solution, which therefore allows infinite turndown in the quantity of sulfur contained in the feed gas. However, the cost of a Stretford plant is largely a function of the amount of solution that must be circulated, and therefore a goal in piloting the process is to determine the maximum solution loading possible that will result in adequate operation of the system while obtaining a satisfactory level of H<sub>2</sub>S removal effectiveness.

The use of the ejector-venturi gas scrubbing system (see Appendix E) affords wide gas turndown capability for the system. Since the flow of gas is motivated by the liquid spray, the scrubbing system functions 3/8"

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largely independently of the gas flow rate. The scrubber, which has a maximum gas capacity of 28.3 sm<sup>3</sup>m, can be turned down to nearly zero. For gases containing larger amounts of sulfur the solution loading can be adjusted by reducing the gas throughput.

By changing the ejector liquid-spray nozzle the rate of liquid circulation can be reduced. Therefore gases with less sulfur can be accommodated while maintaining high solution loadings.

The operation of the pilot plant would approximate that described in Table 40 for various gas feeds.

### Equipment Description

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The pilot plant would be a portable skid-mounted unit, completely assembled with all equipment, controls, and instrumentation required for a complete and operable unit. Utilities and services needed to operate the pilot plant must be supplied at the test site and be field connected.

The maximum demand for services is as follows:

Raw retort gas feed Treated gas return Electrical power Process water Cooling water Sour-water disposal

 $28.3 \text{ sm}^{3}\text{m}$  (30.5cm-diam duct)

 $28.3 \text{ sm}^{3}\text{m}$  (30.5cm-diam duct)

30 hp (460 V, 3-phase, H<sub>2</sub>)

0.11 Lpm

1100 gpm

4.9 Lpm

A list of the major equipment required for construction of the pilot plant is given in Table 41.

#### E. PRELIMINARY COST ESTIMATE

The total estimated cost of the pilot plant with all equipment, instru- in the ART ments, and controls as described, assembled on skid mountings as a complete and operable unit, is as follows:

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Table 40. Approximate Pilot-Plant Operation for Various Gases

Gas	Type of Retort	H <sub>2</sub> S (vol %)	Gas Flow (Gm3m)	Sulfur	Liquid Flow	Solution Loading (ppm wt)
Tosco	Indirect	4.12	2.37	6.6	159	800
Union	Indirect	3.82	2.55	_6.6	159	800
Paraho	Direct	0.30	32.4	6.6	159	800
Geokinetics	Direct	0.13	31.1	4.3	102	518
Оху	Direct	0.119	31.1	4.0	102	476

Table 41, , Pilot-Plant Components

Component (1 Each)	Construction Material	Description
	Item No. 100: Gas-Cooler	Gas-Cooler Module (1)
Gas cooler	Type 304 SS	Complete skid-mounted assembly as follows: packaged column, 76.2cm-diam X 304.8cm straight side, with 122cm of 5.1cm Intalox saddles; vertically mounted; atm pressure, 60°C
Heat exchanger	Type 304 SS shell and tubes	$35.52m^2$ ; horizontally mounted; atm pressure, $60^{\circ}$ to $32^{\circ}$ C.
Circulation pump	Type 316 SS	1136 % pm, 30 psig, 10 hp.
Flow control		<pre>Gas flow control value with flow sensor trans- mitter, controller, and recorder</pre>
Level control		Liquid-level control valve with sensor, controller, and high- and low-level alarms
Temperature		Multiple-point temperature recorder with temperature sensors
Pressure indicator		Direct-reading pressure gage (in inches of water)
Gas analyzer	:	Gas/liquid chromatograph for measuring $_{ m H_2S}$ in gas stream with manual samples
Skid mounting	Steel	As required

Component	Construction Material	Description
(T Bacil)	200:	Absorber Module (1)
Venturi scrubber	FRP with type 304 SS nozzle	Complete skid-mounted assembly as follows: 5.1 cm, suction and discharge; 31.1 am <sup>3</sup> m at 32°C; 1.27 to 2.54cm water gage draft; .59 lpm water solution at 1.2 sp. gr.; 60 psig
Mist eliminator	Stee1	681.3% vertical tank with 66 cmdiam X 15.2cm-deep type 304 SS demister pad
Delay tank	Stee1	2460 %,91.4cmdiam X545 cm straight side; vertical tank
Solution pump Flow control	Malleable iron	189 $\ell$ , 60 psig, 3 hp Liquid-flow control valve with flow sensor, transmitter, controller, and recorder
pH monitor Level control		pH sensor, transmitter, and recorder Liquid-level control valve with sensor and high-level alarm
Temperature sensor		Temperature sensor, connected to temperature recorder for Item No. 100
Skid mounting	Steel	As required

Table 41. (Cont'd)

	Material	Description
	Item No. 300:	Oxidizer Module (1)
Oxidizer tank	Stee1	Complete skid-mounted assembly as follows: 45.7 cmdiam X30.5 cm straight side; vertically mounted, with sludge ring and spout for sulfur slurry; cone hood type of vent top with 366 cm stack
Air blower	Standard	1.42 sm <sup>3</sup> n,15 psig, with inlet air filter
Injector	Type 304 SS	Air, 1.05 sm <sup>3</sup> mat 8 psig; liquid, 159 ± sp. gr. 1.2
Solution pump tank	Steel	4542 &; 122 cm diam X 366-cm straight side; vertically mounted
Flow control		Air flow control valve with flow transmitter controller and recorder
Sight glass		Liquid-flow sight glass
Level indicator		Liquid-level indicator with low-level alarm
Skid mounting	Steel	As required

Table 41: (Cont'd)

Component (1 Each)	Construction Material	Description
	Item No. 400: Sulfur Module (1)	Tur Module (1)
Slurry tank	Steel	Complete skid-mounted assembly as follows: 681- & 76.2cm -diam X 137-cm straight side; open top
Agitator	Steel	Clamp type; mild agitation; 0.25 hp
Slurry pump	Duriron	5.7 %, 45 psig, 0.25 hp
Sulfur filter	Type 304 SS	45.7cm diam X122cm st. side; vertically mounted plate type; 50 psig
Sulfur sluice pump	Duriron	37.8 Apm low head, 0.5 hp
Sulfur transfer box	Steel	2.3m <sup>3</sup> (2271 , 8); closed dumpster
Skid mounting	Stee1	As required

Item-100 gas cooler module	\$200,000
Item-200 absorber module	74,000
Item 300 oxidizer module	68,000
Item 400 sulfur handling module	58,000
Total cost (December 1980 capital)_	\$400,000

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If the gas were provided at 22°C, the gas cooler module could be eliminated, in which case the total cost of the system would be about \$260,000. The cost of the gas cooler module (Item 100) includes the instrumentation and controls required for proper functioning of the pilot unit. These instruments (\$60,000) would be required even if the cooler module were eliminated.

The probable cost range is estimated as follows:

Range	With Cooler	Without Cooler
High	\$520,000	\$338,000
Probable	400,000	260,000
Low	308,000	200,000

### F. ADVANTAGES AND USE OF PILOT PLANT

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The primary purposes of a pilot plant are to prove the technical feasibility of the process application and to obtain data needed for scaleup of the process to production-size equipment. If the performance of the pilot plant both at optimum and extreme conditions matched the predicted results and if there were no unreconcilable problems, then commercial application of the process could be carried out with confidence. The advantages to the EPA of piloting the process would be to

- 1. determine the characteristics of the raw retort gas,
- 2. determine the characteristics of the treated gas,
- 3. determine the characteristics of the waste streams,
- 4. predict the characteristics of controlled emissions,
- 5. determine the process efficiency and reliability,

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- 6. evaluate transient conditions of startup, normal and emergency shutdown, process upsets, etc.,
- 7. evaluate the economics of the Stretford process.

Piloting of the process can aid in commercial application of the process by providing demonstration of continuous operation of the system, as well as the following data:

- Level of H<sub>2</sub>S and other sulfur compounds in the treated gas and percentage of overall removal of sulfur compounds by the Stretford process
- 2. Disposition of COS and other organic sulfur compounds in the feed
- 3. Effects of unsaturated hydrocarbons in the feed gas on process operation and life of the Stretford chemicals
- 4. Amount of hydrocarbons or other contaminants in the oxidizer vent
- 5. Quality of sulfur produced
- 6. Solution sulfur loading versus gas characteristics
- 7. Rate of thiosulfate formation versus gas characteristics
- 8. Absorption of CO2 versus gas characteristics
- 9. Design parameters for the absorber
  - a. HS loading in the solution
  - b. Thiosulfate concentration
  - c. Na<sub>2</sub>CO<sub>3</sub> concentration
- Design parameters for the oxidizer
  - a. Ratio of air to sulfur loading
  - b. Ratio of air to tank diameter
  - c. Height of oxidizer
  - d. Characteristics of sulfur froth
- 11. Design parameters for filter
  - a. |Filter selection
  - b. Pumping rates and pump type
  - c. Filtration rates and wash cycles required

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				JOB NUMBER
PROJECT				9212
STRETFORD DI	RECT PROCESS	ВУ	DATE	EF NUMBER
PHASE			1-31-80	

PRE-1 Case A SECTOR NUMBER & NAME

SECTION NU	MBER & NAME				
-	BASE X ESC. X CAPACITY	KQUANT. X	[MATERIAL +	CONDITION + COMPLEXITY]	== M \$ (19 )
ITEM	NAME OF FACILITY	QUANT.	CONSTRUCTION MATERIAL	DESCRIPTION	· <del></del>
101	Gas Coolers	6	304 St.Stl	12'\Packed section 2000 ft 3 - 3 1/2"\Packed Rings	
102	Cooler Cir. Pumps	6	304 St.Stl	_	
103	Exchangers	6	304 St.Stl	5000 ft <sup>2</sup> , U Tube, 14' LG Tube	
104	Trace & Cover For Freeze Protection			All Items Subtotal (100 Series)	\$4,028
201	Absorbers	16	Steel	12'Φx42' Bl to Bl 18' Packed section 2000 ft - 3 1/2" Φ Pall Rings	
202	Fwd. Pumps	16	FRP	500 GPM, 40' HD, 10 HP	
203	Oxidizer Tower	1	Stl. & Epoxy	30'4x24' HI w/foam Trough & Air Ring	
204	Air Diffusion Sys.	1	Steel	7500 SCFM, Blowers, Elec., Air Piping, Bldg.	
205	Sulfur Slurry Tk.	1	Steel & Epoxy	15000 Gal, 25 PSI	
206	Tank Agitator	1	Steel	Medium Ag., 15 HP.	
207	Slurry Pump	1	FRP.	100 GPM, 40 HD, 3 HP	
208	Rot. Vac. Filter Sys.	i	Stl.	Incls; Filter, Vac. Pump, Filtrate Receiver & Pump, Conveyor, Wash Tk, & Pump, Building	
209	Sulfur Melter	1	Steel	1200 Gal., 25 PSI, JKT. & Internal Coil, Insulate & Baffle	
210	Sulfur Fwd. Pump	1	Steel	10 GPM, 1 HP	
211	Sulfur Store Tk.	1	Steel	30,000 Gal. 1 WK API. S.G. = 2	
212	Agitator	1	Steel	Heavy, 5 HP	

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PROJECT				1	9212
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	BASE X ESC. X CAPACITY	KQUANT.	( [MATERIAL -	CONDITION + COMPLEXITY]	== MS(19 )
FACTORS	NAME OF FACILITY	QUANT.	CONSTRUCTION MATERIAL	DESCRIPTION	
213	Drier Sys.	1	Steel	200 ft <sup>2</sup> W/Heating, Air or Vac. Sys., Feed & Disch. Conveyor, Bldg.	. '
214	Make-up Mix Tk.	1	Steel	2700 Gal, 8 Hr. Store	
215	Fwd. Pump	1	st.stl.	100 GPM, 100' HD, 7 1/2 HP	
216	Lean Sol. Fd. Tk.	1	Steel Epoxy	36,000 Gal, 5 min. SG - 1.2	·
217	Tower Cool Pumps	8	FRP	1000 GPM, 80'_ HD, 30_HP	
218	Purge Pump	1	FRP	100 GPM, 40' HD., 3 HP	·
219	Mix Tk. Agitator	1	Steel	Medium, 25 HP	,
220	Trace & Cover for Freeze Protection			All Items As Required	
				Subtotal (200 Series)	\$5943
				Total (All Equipment)	\$9971
,					

### PRELIMINARY CAPITAL - DETAIL SHEET OF PROJECT PHASE PRE-1 · Case A SECTOR NUMBER & NAME SECTION NUMBER & NAME CONDITION + COMPLEXITY] = MS(19)[MATERIAL BASE X ESC. X CAPACITY X QUANT. X FACTORS CONSTRUCTION DESCRIPTION QUANT. ITEM NAME OF FACILITY MATERIAL Case Al - Disposal of Purge Solution \$ 9,971 Equipment as above (Sheets 1 and 2) 1,994 Allowance (20%) 11,965 Total 1,197 Royalty (10% of capital) 160 Initial chemical charge 1980 Installed Capital Cost \$13,322 Case A2 - Recovery and Recycle of Purge Solution Equipment as above (Sheets 1 and 2) \$ 9,971 Purge stream reductive incineration system 1,150 (300 lb/hr salts) Sub-total 11,121 2,224 Allowance (20%) \$13,345 Total 1,335 Royalty (10% of capital) Initial chemical charge 160 1980 Installed Capital Cost \$14,840

#### SHEET PRELIMINARY CAPITAL - DETAIL SHEET OF 6 9212 STRETFORD DIRECT PROCESS EF NUMBER DATE PHASE Case Al PRE-1 SECTOR NUMBER & NAME SECTION NUMBER & NAME = MS(19)CONDITION + COMPLEXITY] X QUANT. X **[MATERIAL** BASE X ESC. X CAPACITY FACTORS CONSTRUCTION DESCRIPTION QUANT. NAME OF FACILITY ITEM Disposal of Purge Solution Raw Materials: 1. 174,000/yr Vanadate 887,000/yr ADA 24,000/yr EDTA. 3,000/yr Sodium Carbonate Neg/yr Iron (Soluble) \$1,088,000/yr By Products 2. 12028 tons/yr x \$90/ton = \$1,083,000 (carbon) Sulfur Water Treatment or Disposal 3. 132x10<sup>6</sup> gal/yr 9.28x10 lbs/yr 5x10 gal/yr No charge "Sour" Water No charge "Purge" Water No charge "Filter Wash" Utilities 4. 10.1x10<sup>6</sup> KwH/yr \$303,000 Electrical Power 28x10<sup>6</sup> lb/yr 70,000 150 Psig Steam 526,000 Process & Cooling Water 10,000 GPM \$899,000 5. Manpower \$420,000 14 men x 2000 hr/yr x \$.15/MnHr 180,000 1 supervisor 8760 hr x \$20/MnHr \$600,000 Maintenance, (5%), Capital Rec. (20%), Misc. (4%) 6. \$3,864,000/yr 29% x capital \$13,322,000

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	DACE.	X ESC. X CAPACITY	X QUANT. )	( [MATERIAL +	CONDITION +	COMPLEXITY]		MS(19
CTORS			QUANT.	CONSTRUCTION MATERIAL	DESCRIP	<del></del>		
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			<b>-</b>	•				
_					\$5,368,000/yı	-		ŧ
7.	Net: Ani	nualized Cost			\$0.294/ton of			
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#### PRELIMINARY CAPITAL - DETAIL SHEET 6 OF STRETFORD DIRECT PROCESS EF NUMBER DATE PHASE Case A2 PRE-1 SECTOR NUMBER & NAME SECTION NUMBER & NAME = MS(19)CONDITION + COMPLEXITY] **MATERIAL** X QUANT. X BASE X ESC. X CAPACITY FACTORS CONSTRUCTION DESCRIPTION QUANT. NAME OF FACILITY MATERIAL ITEM Recovery and Recycle of Purge Solution 1. Raw Materials: \$887,000/yr ADA 24,000/yr EDTA \$911,000/yr By Products: 2. \$1,116,000/yr (Credit) Sulfur, 12400 TPY Water Treatment or Disposal 3. 132x10<sup>6</sup> gal/yr No charge "Sour" Water 5x10<sup>6</sup> gal/yr No charge "Filter Wash" Utilities: 4. 11.5x10 KwH/yr 345,000/yr Elec. 42.9x10 lb/yr 107,000/yr Sth. 578,000/yr 11000 Gpm Water 328,000/yr 15 M Btu/hr Gas \$1,358,000/yr 5. Manpower \$540,000/yr 18 men x 2000 hr/yr x \$15/MnHr 180,000/yr 1 super. 8760 hr/yr x \$20/MnHr \$720,000/yr . .... Maint. (5%), Capital Rec. (20%), Misc. (4%) \$4,304,000/yr 6. \$6,177,000/yr Net Annualized Cost 7. \$0.338/ton of shale \$0.569/bbl of oil Total Installed Capital 8. \$19,292,000 Dec. 1980 14,840,000 Dec. 1980 Probable

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TAIL GAS TREATMENT

PHASE

PRE-1 Case B

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9212

EF NUMBER

FF NUMBER

Case B

SECTION NUMBER & NAME SECTOR NUMBER & NAME

SECTION NU	MBER & NAME	SECTOR NO	MBEK & NAME		
FACTORS	BASE X ESC. X CAPACITY	Y QUANT. Y	( [MATERIAL +	CONDITION + COMPLEXITY]	== MS(19 )
ITEM	NAME OF FACILITY	QUANT.	CONSTRUCTION MATERIAL	DESCRIPTION	
101	Gas Cooler	6	304 St.Stl.	12'0x42' B1-B1, 10 PSI, 18' of Packing, 2000 ft	
102	Cooler Pump	6	304 St.Stl.	1000 GPM, 100' HD, 40 HP	
103	Exchangers	6	304 St.Stl.	5000 ft <sup>2</sup> U Tube, 14' Lg.	
104	Trace & Cover For Freeze Protection			All Items Subtotal (100 Series)	\$4,028
201	lst. Stage Absorber	16	Steel	12'Φx42' Bl-Bl, 10 Trays	
202	Exchanger	16	Steel	829 ft <sup>2</sup>	
203	Pump	16	D.I.	150 GPM, 100' HD, 7 1/2 HP	
204	Interchanger	16	Steel	287 ft	
205	lst Stage Desorber	3	Steel	12'Φ42' Bl-Bl, 10 Trays	
206	Condenser	3	Steel	3000 ft <sup>2</sup> , Float, 14' Lg	
207	Pump	3	D.I	716 GPM, 150' HD, 50 HP	
208	Pump	3	D.I.	639 GPM, 60' HD, 15 HP	
209	Reboiler	3	Steel	820 ft <sup>2</sup>	
210	Compressor	3	Std.	6200 ACFM, 2 PSI, 25 HP	1
211	Amine Recovery Reboil	1	Steel	2300 Gal., 154 ft <sup>2</sup>	·
212	Sludge Pump	1	D.I.	1/2 HP	
213	Sludge Accumulator	1	Steel	10 Day, 150 Gal	
214	Trace & Cover For Freeze Protection			All Items As Required Subtotal (200 Series)	\$6,205
		· ·			1
301	2nd Stg. Absorber	1	Steel	9'4x36' Bl-Bl, 10 Trays	
302	Pump	1	D.I.	506 GPM, 150' HD, 25 HP	
303	Interchanger	1	Steel	700 ft <sup>2</sup>	
304	Cooler	1	Steel	2200 ft <sup>2</sup>	
<u> </u>					

SECTION NUMBER & NAME

SHEET .

PROJECT 3 STAGE SELECTIVE ABSORPTION PLUS CLAUS SULFUR RECOVERY WITH SCOT

TATI. GAS TREATMENT

PHASE

PASE

BY

DATE

JOB NUMBER

9212

EF NUMBER

Case B | |

		<u> </u>			
FACTORS	BASE X ESC. X CAPACITY	X QUANT.	( [MATERIAL +	CONDITION + COMPLEXITY	= MS(19 )
ITEM	NAME OF FACILITY	QUANT.	CONSTRUCTION MATERIAL	DESCRIPTION	
305	2nd Stg. Desorber	1	Steel	9'Φx36' Bl-Bl, 10 Trays	:
306	Pump	1	D.I.	432 GPM, 150' HD, 40 HP	
307	Pump	1	D.I.	432 GPM, 150' HD, 40 HP	
308	Reboiler	1	Steel	900 ft <sup>2</sup>	·
309	Condenser	1	Steel	2000 ft <sup>2</sup>	
310	Compressor	1	Std.	6200 ACFM, 2 PSI, 25 HP	
311	Trace & Cover For Freeze Protection			All Items As Required Subtotal (300 Series)	\$957
401	3rd Stg. Absorber	1	Steel	6'Φx30' Bl-Bl, 10 Trays	
402	Pump	1	D.I.	200 GPM, 100', 20 HP	
403	Interchanger	1	Steel	187 ft <sup>2</sup>	
404	Cooler	1	Steel	823 ft <sup>2</sup>	
405	3rd Stg. Desorber	1	Steel	6'4x30' Bl-Bl, 10 Trays	
406	Pump	1	D.I.	160 GPM, 150' HD, 15 HP	: .
407	Pump	1	D.I.	160 GPM, 150' HD, 15 HP	
408	Reboiler	1	Steel	300 ft <sup>2</sup>	
409	Condenser	1	Steel	731 ft <sup>2</sup>	
410	Compressor	1	Steel	2123 CFM, 7 1/2 HP	
411	Trace & Cover For Freeze Protection			All Items As Required Subtotal (Series 400)	\$462
500	3 Stage Claus Unit Trace & Cover As Required	Sys.		2123 scfm, 27.4 mole % H <sub>2</sub> S 35.8 TPD of Sulfur	\$566
•-	-				

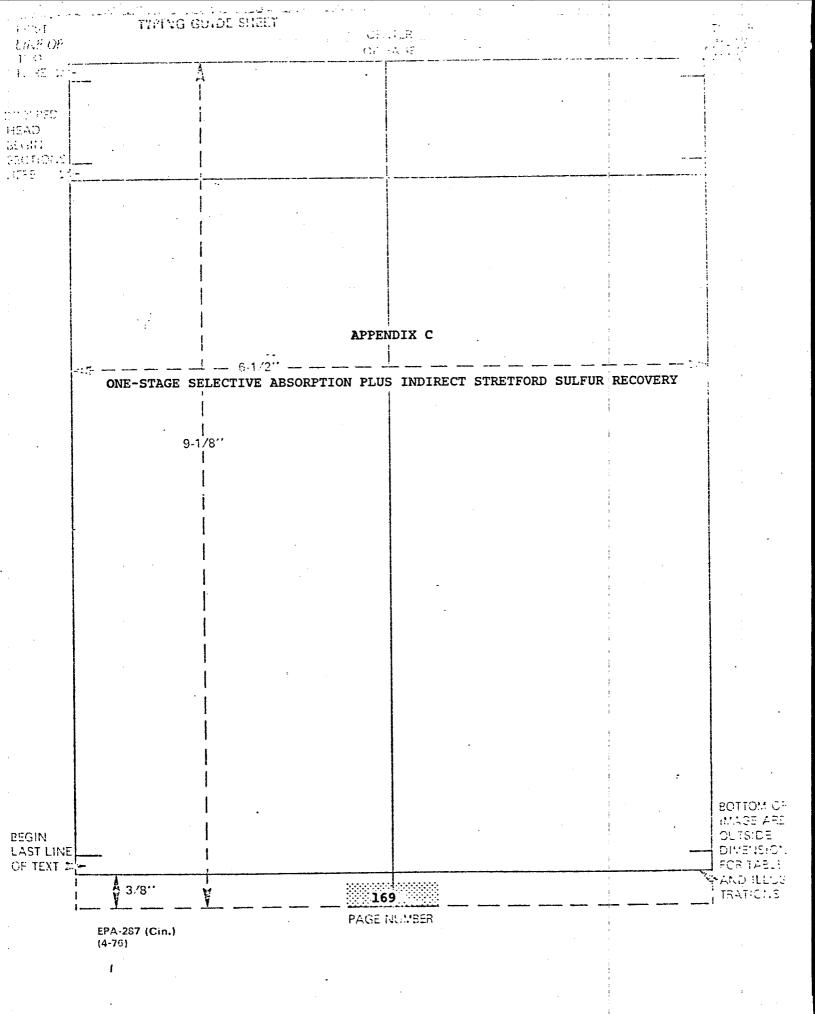
SHEET

										OF	4
PROJECT	3 STAGE	SELECTIVE	ABSORPTION	PLUS CLAU	S SULFUR	RECOVERY WIT	TH SCOT		JOB NUMBER		
		S TREATMEN		1200 02.0	·				9212		
PHASE	<u> </u>	CASE	BY			DATE		,	EF NUMBER		
PRE-1		Case B		•				į			
SECTION NUMBI		Case_b	l:	ECTOR NUMBER &	NAME						

SECTION NU	MBER & NAME	SECTOR NU	MBER & NAME		k.
FACTORS	BASE X ESC. X CAPACITY	X QUANT. X	[MATERIAL +	- CONDITION + COMPLEXITY]	= MS(19 )
ITEM	NAME OF FACILITY	QUANT.	CONSTRUCTION MATERIAL	DESCRIPTION	
500	SCOT tail Gas Treatment System Trace & Cover As Required	Sys.		3225 scfm, 0.035 mole % H <sub>2</sub> S 164 bb/day of sulfur	\$ 358
	· ;·			Total (All Equipment)	\$12,576
	Allowance (20%)		•		\$ 2,515
-	Royalty (Per Dow Chemical U	A)			300
	Initial Chemical Charge	-			160
		1980	Installed (	apital Cost	\$15,551
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4 OF

3 STAGE SELECTIVE ABSORPTION PLUS CLAUS SULFUR RECOVERY WITH SCOT PROJECT 9212 TAIL GAS TREATMENT EF NUMBER Case B SECTOR NUMBER & NAME SECTION NUMBER & NAME = MS(19)CONDITION + COMPLEXITY] XQUANT. X **[MATERIAL** BASE X ESC. X CAPACITY FACTORS CONSTRUCTION DESCRIPTION QUANT. NAME OF FACILITY ITEM MATERIAL 1. Raw Materials: \$ 24,000/yr 13.14 tons/yr MDEA 2. By Products: \$1,171,000/yr (Credit) 13,016 Tons/yr Sulfur To Water Treatment or Disposal: 3. 133.4x10<sup>6</sup> gal/yr 3x10<sup>4</sup> lb/yr No charge "Sour" Water No charge "Sludge" Water Utilities: 4. 2873x10 KwH/yr \$ 158,000/yr Elec. Power 7,183,000/yr Steam, Net 1.5x10 gal/yr 1,545,000/yr Cool Water \$8,886,000/yr 5. Manpower: \$300,000/yr 10 Operators 180,000/yr 1 Supervisor \$480,000/yr Maintenance (5%), Capital Rec. (20%), Misc. (4%) 6. \$4,510,000/yr 20% x Capital \$15,551,000 \$12,729,000/yr Net Annualized Cost 7. \$0.069/Ton of Shale \$1.172/bbl of Oil Total Installed Capital 8. \$20,216,000 High, Dec. 1980 15,551,000 Prob., Dec. 1980 11,974,000 Low, Dec. 1980



SHEET

JOB NUMBER PROJECT 9212 EF NUPBER I STAGE SELECTIVE ABSORPTION PLUS INDIRECT STRETFORD SULFUR PHASE CASE BY 2-4-80 C

PRE-1 SECTOR NUMBER & NAME

SECTION NUA	MBER & NAME				40
	BASE X ESC. X CAPACITY	XQUANT.	( [MATERIAL +	CONDITION + COMPLEXITY]	= MS(19)
FACTORS	NAME OF FACILITY	QUANT.	CONSTRUCTION MATERIAL	DESCRIPTION	
101	Gas Coolers	6	304 st.stl.	12' \$\Partial x 42' Bl-Bl, 10 PSI, 18' Pack Sec. 2000 ft 3 1/2" Pall Rings	
102	Cooler Pumps	6	304 st.stl.	1000 GPM, 100' HD, 40 HP	
103	Exchangers	6	304 st.stl.	5000 ft <sup>2</sup> , U Tube, 14' LG	
104	Trace & Cover For Freeze Protection			All Items Subtotal (100 Series)	\$4,028
201	Selective Absorber	16	Steel	12' 4x42' Bl to Bl 10 Valve Trays	
202	Exchanger	16	Steel	57/ft <sup>2</sup>	
203	Pump	16	p.1. "	150 GPM, 100' HD, 7 1/2 HP	
204	· Interchanger	16	Steel	287 ft <sup>2</sup>	
205	Strip Column	3	Steel	12'4x42' Bl-Bl 10 Plate	,
206	Condenser	3	Steel	3000 ft <sup>2</sup> , Float, 14' LG	
207	Column Pump	3	D.1.	716 GPM, 150' HD, 50 HP	
208	Reboil Pump	3	D.1.	639 GPM, 50' HD, 15 HP	
209	Strip Col. Reboiler	3	Steel	722 ft <sup>2</sup>	
210	Compressor	3	std.	6200 ACFM, 2 PSI	
211	Amine Recv. Reboil	1	Steel	2300 Gal., 154 ft <sup>2</sup>	
212	Sludge Pump	1	D.1.	1/2 HP	
213	Sludge Accumulator	1	Steel	10 Day, 150 gal.	
214	Trace & Cover For Freeze Protection			All Items As Required Subtotal (200 Series)	\$5,957
301	H <sub>2</sub> S Absorber	1	Steel	10'x28' Bl-Bl, 13' Pack, Sec 2200 ft <sup>3</sup> -3 1/2" Pall Rings	•
302 <sup>-</sup>	Oxidizer Tower	1	Steel Epoxy	30'4x24' W/Foam Trough Air Ring	!

2 OF 3

PROJECT								1	JOB NUMBER	
T STAGE	SELECTIVE	ABSORPTION	PLUS	INDIRECT	STRETFORD	SULFUR	RECOVERY	1	9212	
PHASE	CASE		BY				DATE		EF NUMBER	
PRE-1		Ç					2-4-80	: "		
SECTION NUMBER	R & NAME	,		SECTOR NUMBE	R & NAME			i		

PRE-1	<u> </u>	ISECTOR NU	MBER & NAME	2-4-80	
SECTION NU	MBER & NAME	SEC. OK III			_
	BASE X ESC. X CAPACITY	X QUANT. >	( [MATERIAL +	CONDITION + COMPLEXITY]	= MS(19 )
FACTORS	NAME OF FACILITY	QUANT.	CONSTRUCTION MATERIAL	DESCRIPTION	
		1	Steel	7500 SCFM, Blowers, Elec.	
303	Air Diffusion System			Air Pipe, Bldg.	
304	Sulfur Slurry Tk.	1	Steel Epoxy	15000 Gal. 25 psi	
305	Tank Agitator	1	Steel	Med., 15 HP	
306	Slurry Pump	. 1	FRP	100 GPM, 40' HD, 3 HP	,
307	Rot. Vac. Filter	1	Std.	Filter, Vac Pump, Filtrate Receiver & Pump, Conveyor,	
-		·		Wash Tk. & Pump, Bldg.	
308	Sulfur Melter	1	Steel	1200 Gal., 25 Psi, JKT. & Internal Coil, Insulate & Baffle	•
					<del></del>
; 309	Sulfur Fwd. Pump	1	Steel	10 GPM, 1 HP, T&C	,
310	Sulfur Store Tk.	1	Steel	30,000 Gal. T&C, 1 wk API s.g. = 2	
311	Agitator	1	Steel	Heavy, 5 HP	
312	Drier Sys	1	Steel	200 ft <sup>2</sup> w/Heating, Air or Vac. Sys., Feed & Disch. Conveyor, Bldg.	
313	Make-up Mix. Tk.	1	Steel	27,000 Gal., 8 Hr Store	
314	Fwd. Pump	1	st.stl.	100 GPM, 100' HD, 7 1/2 HP	
315	Lean Sol. Fd. Tk.	1	Steel Epoxy	36,000 Gal. 5 min., s.g.=1.2	
316	Purge Pump	1	FRP	100 GPM, 40; HD., 3 HP	
317	Mix Tk. Agitator	1	Steel	Medium, 25 HP	,
318	Trace & Cover For Freeze Protection			All Items As Required	
				Subtotal (300 Series)	\$ 2,924
				Total (All Equipment)	\$12,909
	•			*	

### PRELIMINARY CAPITAL - DETAIL SHEET JOB NUMBER PROJECT 9212 I STAGE SELECTIVE ABSORPTION PLUS INDIRECT STRETFORD SULFUR RECOVERY EF NUMBER 2-4-80 C PRE-1 SECTOR NUMBER & NAME SECTION NUMBER & NAME CONDITION + COMPLEXITY = M \$ (19)[MATERIAL X QUANT. X BASE X ESC. X CAPACITY FACTORS CONSTRUCTION DESCRIPTION QUANT. NAME OF FACILITY ITEM MATERIAL Case Bl - Disposal of Stretford Purge Solution \$12,909 Equipment as above (sheets 1-3) 2,582 Allowance (20%) \$15,491 Total 1,549 Royalty (10% of capital) 160 Initial Chemical Charge \$17,200 1980 Installed Capital Cost Case B2 - Recovery and Recycle of Stretford Purge Solution \$12,909 Equipment as above (sheets 1-3) 1,150 Stretford Purge Stream Reductive Incineration System \$14,059 Subtotal 2,812 Allowance (20%) \$16,871 Total 1,687 Royalty (10% of capital) 160 Initial Chemical Charge \$18,718 1980 Installed Capital Cost

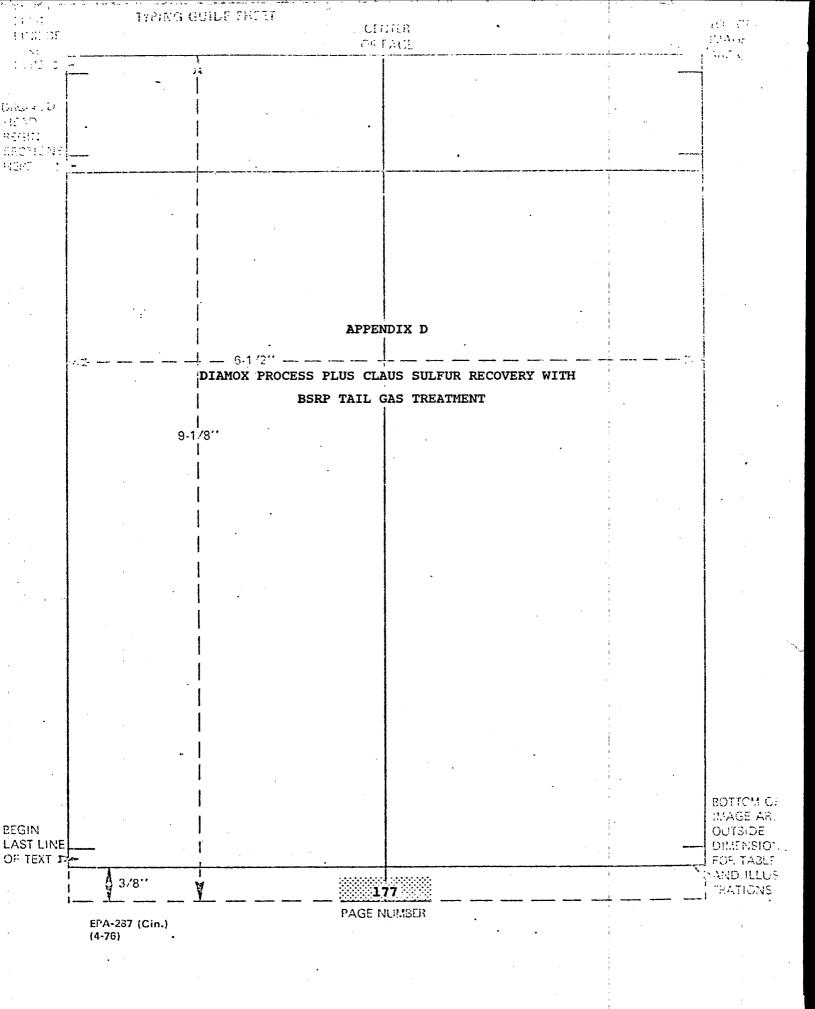
1 of 4

OJECT				JOB NUMBER
I STAG	E SELECTIVE ABSORPTI	ON PLUS INDIRECT STRETFORD	SULFUR RECOVERY	9212 EF NUMBER
SE	CASE	BY	DATE	EP NUMBER
	Cl	SECTOR NUMBER & NAME		<u> </u>
TION NUM	BER & NAME	SECTOR NUMBER & NAME	:	
			+ CONDITION + COMPLEX	(ITY] = MS(19)
TORS	BASE X ESC. X CAPACIT	Y XQUANT. X [MATERIAL -		
EM	NAME OF FACILI	TY QUANT. CONSTRUCTION	DESCRIPTION	
			· 1	
		•		
Disp	osal of Purge Soluti	on		
			· · · · · · · · · · · · · · · · · · ·	
_			† !	
1.	Raw Materials:			
	Methyl Diethanola	mine	\$ 28,000/yr	
	ADA		954,000/yr 24,000/yr	
	EDTA		181,000/yr	
	Vanadate		Neg/yr	
	Sodium Carbonate		3,000/yr	i i
	Iron	•		
	•		\$1,190,000/yr	
			1	
_	Des Des des cha			
2.	By Products	<u>.</u>	63 330 000 6	+ (Cmoditi
	Sulfur 12,516	5 ton/yr	\$1,130,000/yi	Crearing
	•			
3.	To Water Treatment	or Disposal:	:	
٥.		6	No charge	
	"Sour" Water	132x10 <sup>6</sup> gm/yr 9.65x10 1b/yr	No charge	
	"Purge" Water	9.65×10 1b/yr	No charge	
	"Sludge" Water "Filter Wash"	3x10 <sup>4</sup> lb/yr 5x10 <sup>6</sup> gal/yr	No charge	
	"Filter Wasn"	5X10 gai/yi	210 02122 32	
	_	•		
4.	Utilities:			
	Electrical Power	7.8x10 <sup>6</sup> KwH/yr 1525x10 kb/yr	\$ 234,000/y:	r i
	150 Psig Steam	1525x10 <sup>6</sup> kb/yr	<b>3,812,0</b> 00/y:	r .
	Water	24M GPM	1,261,000/y	<u>r</u>
			\$5,307,000/y	r
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ELIM	INARY CAPIT	AL - DET	AIL SHEET			JOB NUMB	SHEET  2 OF 4  ER
JECT					THE PRINTER A POTON		9212
I ST	AGE SELECTIVE A	SORPTION PLU	JS INDIRECT STR	ETFORD DES	DATE	EF NUMB	R
)E	C1_						
ION NUME	BER & NAME		SECTOR NUMBER & NAME				
					CONDITION + COMPLEXI	tv]	= MS(19
TORS	BASE X ESC. X C	APACITY	XQUANT. X [MATERIA				- 1115(13
EM	NAME OF F	CILITY	QUANT. CONSTRUCT	AL	DESCRIPTION		
5.	Manpower  15 men x 200  1 supervisor		5/yr Rec. (20%), Mis	sc. (4%)	\$450,000/yr 180,000/yr \$630,000/yr		
6.	29% x \$17,20		Rec. (200), 122	(10)	\$4,988,000/yr		
7.	Net Annualized	Cost			\$10,985,000/y \$0.602/ton of \$1.011/bbl of	shale	
	Total Installe	d Capital					}
8.	High, Dec. 1 Probable, Dec. 19	.980 c. 1980		·	\$22,360,000 17,200,000 13,244,000		
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#### SHEET PRELIMINARY CAPITAL - DETAIL SHEET 3 aF JOB NUMBER PROJECT I STAGE SELECTIVE ABSORPTION PLUS INDIRECT STRETFORD DESULFURIZATION EF NUMBER C2 SECTOR NUMBER & NAME SECTION NUMBER & NAME CONDITION + COMPLEXITY] == MS(19 X QUANT. X **MATERIAL** BASE X ESC. X CAPACITY FACTORS CONSTRUCTION MATERIAL DESCRIPTION QUANT. NAME OF FACILITY ITEM Raw Materials: 1. 28,000/yr Methyl Diethanolamine 954,000/yr ADA 24,000/yr EDTA. \$1,006,000/yr 2. By Products \$1,160,000/yr (Credit) 12,903 TPY x \$90/T Sulfur Watertreat or Disposal 3. 133.4x10<sup>6</sup> gal/yr 3x10<sup>4</sup> 1b/yr 5x10<sup>6</sup> gal/yr No charge "Sour" Water No charge "Sludge" Water No charge "Filter Wash" Utilities: 9.2x10<sup>6</sup> KwH/yr 1540x10<sup>6</sup> lb/yr \$ 276,000/yr Elec. Power 3,850,000/yr Steam 1,314,000/yr 25M GPM Water 328,000/yr 15M Btu/hr Gas \$5,768,000/yr 5. Manpower \$570,000/yr 19 men x 2000 hr/yr x \$15/hr 180,000/yr 1 supervisor \$750,000/yr Maintenance (5%), Capital Rec. (20%), Misc. (4%) 6. \$5,428,000/yr 29% x \$18,718,000

		CAPITAL -					DTEX (0T 037		108 NUMBE 92	
ST	AGE SELECT	TIVE ABSORPT	ION PLU	S INDIF	ECT STRETFORD I	DESULFU  PA	RIZATION	i	EF NUMBER	R
-	CASI	C2	BY					• !		
N NUME	ER & NAME			SECTOR NUM	BER & NAME	•		:		
.,				<u> </u>	F. ATERNAL &	CO	NDITION +	COMPLEXIT	Y] ,	= MS(1
ORS		ESC. X CAPACITY		XQUANT. X	CONSTRUCTION +		DESCRIP			
M	MAM	E OF FACILITY	Y	QUANT.	CONSTRUCTION MATERIAL					
		•					•	1		
			•		· ·		\$11,792	000 /22		
7.	Net Annu	alized Cost					\$0.646/	ton of	shale	
• -					•		\$1.086/	bbl of	oil	
		•						- :		}
	·	;					-	<b>!</b> :		1
8.		stalled Cap	ıtaı:				\$24,333	,000		
	High,	Dec. 1980 ble, Dec. 19	80				\$18,718	,000		
	Low. I	Dec. 1980					\$14,413	,000	•	
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1 OF PROJECT PLUS CLAUS SULFUR RECOVERY WITH BSRP TAIL GAS TREATMENT EF NUMBER Case D SECTION NUMBER & NAME SECTOR NUMBER & NAME X QUANT. X CONDITION + COMPLEXITY BASE X ESC. X CAPACITY [MATERIAL = MS(19)FACTORS CONSTRUCTION QUANT. DESCRIPTION ITEM NAME OF FACILITY MATERIAL 101 Gas Coolers 304 St.St1. 12'Φx42' Bl to Bl, 10 psi 18' packed section, 2000 ft<sup>3</sup>-3 1/2"Φ Pall Rings 102 Cooler Circ. Pumps 1000 gpm, 100" Hd, 40 HP 6 304 St.Stl. 5000 ft<sup>2</sup>, U Tube, 14' LG Tubes 103 Exchangers 6 304 St.Stl. 104 Trace & Cover All Items For Freeze Protection \$4,028 204 Absorbers 6 Steel 12'Φx72' High w/10' skirt, 6 sets spray nozzles deliver 3,000 gpm; 6 sets bulkheads and overflow wiers for liquor collection and visors for gas flow. 202 Absorber Pumps 30 D.I. 3000 gpm pumps to deliver liquor to spray nozzles at 40 psig pressure 203 Stripper Feed Pumps 3000 gpm pumps to deliver D.I. liquor to strippers at 30 psig pressure 204 Stripper Bottoms Pumps 4 D.I. 4500 gpm pumps @ 30 psig pressure 205 Acid Gas Strippers Steel 12' Px7 trays @ 24" spacing 206 Heat Interchanger Steel 18,000 sq.ft. multipass, cross flow 207 Cooling Water Exchanger 4 Steel 18,000 sq.ft. multipass, cross flow 208 2200 sq. ft., 40°F chilled Refrig. Exchanger 4 Steel

Absorber Liquor

Absorber Liquor Storage

Purge Pump

209

210

Steel

CI

4

1

water

180 gpm @ 30 ft. head

540,000 gal., 48'4x40'h

SHEET

JOB NUMBER PROJECT DIAMOX PROCESS PLUS CLAUS SULFUR RECOVERY WITH BSRP TAIL GAS TREATMENT
PHASE | CASE | BY | DATE 9212 EF NUMBER Case D ŠĒ

ECTION NUMBER & NAME	 SECTOR NUMBER & N	AME
	T T T T T T T T T T T T T T T T T T T	

BASE X ESC. X CAPACITY  NAME OF FACILITY  OUANT. CONSTRUCTION MATERIAL  DESCRIPTION  OUANT. CONSTRUCTION MATERIAL  DESCRIPTION  OUANT. CONSTRUCTION MATERIAL  DESCRIPTION  OUANT. CONSTRUCTION MATERIAL  OUANT. OUANT.  OUANT. CONSTRUCTION MATERIAL  OUANT. OUANT.  OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.  OUANT. OUANT.	ON	= MS(19 )
Absorber Feed Pump  Chilled Water Refrig. System  Trace & Cover for Freeze Protection  3000 gpm pumps @ head  1 Std. 1250 Tons, 0°C  All Items As Required Subtotal (200 Ser  300 3 Stage Claus Unit Trace & Cover as Required  400 BSPR Tail Gas Treatment Unit Trace & Cover as Required  Allowance  (20%)  Total  Royalty Initial Chemical Charge (BSPR Unit)		. :
Chilled Water Refrig. System  213 Trace & Cover for Freeze Protection  300 3 Stage Claus Unit Trace & Cover as Required  400 BSPR Tail Gas Treatment Unit Trace & Cover as Required  Allowance  (20%)  Allowance  (20%)  Total  Royalty Initial Chemical Charge (BSPR Unit)	50 psig	
Trace & Cover for Freeze Protection  300 3 Stage Claus Unit Trace & Cover as Required  400 BSPR Tail Gas Treatment Unit Trace & Cover as Required  All Items As Required Subtotal (200 Ser.)  5ys. 10,045 scfm, 6 mo 35.2 TPD of Sulfu 35.2 TPD of Sulfu 427 lb/day  Treatment Unit Trace & Cover as Required  Allowance (20%)  Total  Royalty (Per R. M. Parsons Co.)  Initial Chemical Charge (BSPR Unit)	1 1	
Freeze Protection  300 3 Stage Claus Unit Trace & Cover as Required  400 BSPR Tail Gas Treatment Unit Trace & Cover as Required  Allowance  (20%)  Total  Royalty Initial Chemical Charge (BSPR Unit)  As Required Subtotal (200 Ser  10,045 scfm, 6 mo 35.2 TPD of Sulfu 35.2 TPD of Sulfu 35.2 TPD of Sulfu 35.2 TPD of Sulfu 35.2 TPD of Sulfu 35.2 TPD of Sulfu (20%)  Total  (Per R. M. Parsons Co.)	* i. *	
Trace & Cover as Required  BSPR Tail Gas Treatment Unit Trace & Cover as Required  Allowance  Royalty Initial Chemical Charge (BSPR Unit)  35.2 TPD of Sulful 11,137 scfm, 0.03 427 lb/day Total (20%)  Total  (Per R. M. Parsons Co.)	ies)	\$19,769
Treatment Unit Trace & Cover as Required  Allowance  (20%)  Total  Royalty  (Per R. M. Parsons Co.)  Initial Chemical Charge (BSPR Unit)	le % H <sub>2</sub> S	\$ 1,647
Allowance (20%)  Total (Equipment of the property of the property (Per R. M. Parsons Co.)  Initial Chemical Charge (BSPR Unit)	l mole % H <sub>2</sub> S	\$ 1,021
Royalty (Per R. M. Parsons Co.)  Initial Chemical Charge (BSPR Unit)	pment)	\$26,465
Royalty (Per R. M. Parsons Co.)  Initial Chemical Charge (BSPR Unit)		5,293
Initial Chemical Charge (BSPR Unit)		\$31,758
	<u>.</u>	900
1980 Installed Capital Cost		83
		\$32,741
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#### PRELIMINARY CAPITAL - DETAIL SHEET 3 OF JOB NUMBER PROJECT DIAMOX ABSORPTION PLUS CLAUS SULFUR RECOVERY WITH BSRP TAIL GAS TREATMENT 9212 EF NUMBER Case D SECTION NUMBER & NAME SECTOR NUMBER & NAME CONDITION + COMPLEXITY] = MS(19)[MATERIAL FACTORS BASE X ESC. X CAPACITY X QUANT. X CONSTRUCTION NAME OF FACILITY QUANT. DESCRIPTION ITEM MATERIAL

2. By Products

1.

Raw Materials:

Consumption

Sulfur 12830 TPY

\$1,150,000/yr (Credit)

3. Water Treatment or Disposal:

148 x 10<sup>6</sup> gal/yr

No charge

\$82,680 \$84,000/yr

4. Utilities:

Electric Power	53.9x106 KwH/yr
Steam, Net	4480x10 <sup>6</sup> lb/yr 1.95x10 gal/yr
Cool H <sub>2</sub> O	1.95x10 <sup>10</sup> gal/yr

(Parsons' Info)

\$230/day x 365

Catalysts, Chemicals, Initial Charge

\$ 1,618,000/yr \$11,530,000/yr \$ 1,950,000/yr

\$15,098,000/yr

5. Manpower:

10	Operators
1	Supervisor

\$300,000/yr 180,000/yr \$480,000/yr

6. Maintenance (5%), Capital Rec. (20%), Misc. (4%)

29% x capital \$32,741,000

\$9,945,000

7. Net Annualized Cost

\$24,007,000/yr \$1.315/ton of shale \$2.210/bbl of oil

8. Total Installed Capital

High, Dec. 1980 Prob., Dec. 1980 Low, Dec. 1980 \$42,563,000 \$32,741,000 \$25,211,000