THE CLEVELAND CLIFFS IRON COMPANY One Mining Department ANNUAL REPORT OF GENERAL MANAGER For Year Ending December 31, 1955



#2027

Manager's Annual Report Year 1956

INDEX

	the second s	the second se
Ishpeming District Cliffs Shaft Mine Humboldt Mine Mather Mine "A" Shaft Morris Mine Ohio Mine	192	Page No. 1-17 18-42 43-71 72-78 79-90
Republic Mine Tilden Mine	\ Y x	91-121 122-131
Negaunee District Bunker Hill Mine Cambria-Jackson Mine Maas Mine Mather Mine "B" Shaft Pelletizing Plant	15	132-151 152-170 171-188 189-216 217-253
<u>Tron River District</u> Spies Mine		254
Mesaba District Agnew Mine Alworth Mine Canisteo Mine Cushing Mine Hawkins Mine Hill-Trumbull Mine Holman-Cliffs Mine Sally Mine Sargent Open Pit Wanless Mine		255-262 263-269 270-290 291-292 293-312 313-335 336-356 357-361 362-372 373-380

Continued -



Manager's Annual Report Year 1956

INDEX -2-

	Page No.
Safety Department	With the second s
a. Fatal accidents	381-385
b. Non-fatal accidents	386-401
c. Safety inspection	402-416
d. Ventilation	417-425
e. Mine safety & mine rescue courses	426-428
f. Miscellaneous	429-432
Mining Engineering Department	
a. List of annual report map books	433
b. Map reports	433-435
c. Mining leases	435-439
d. Engineering force	440-446
e. Distribution of time	447-449
f. Costs	449
h. Automobiles	450
i. Mines	450-456
j. Miscellaneous	456-461
Research Laboratory	
I. General Testing Programs	462-472
II. Pyrolysis & agglomeration	473-475
III. Research & development work and flotation projects	476-479
IV. Microscopy section	480-481
V. Fluosolids Reactor Pilot Plant	482-483
VI. Check sampling program	484
The Electric Power Department	
General operations	485-489
Statistical data	490-492
Charts	493
	The second se

Continued -

Manager's Annual Report Year 1956

INDEX -3-

	Page No.
Welfare Department	and the second second second second
General	494-495
1-a. Workman's compensation	496-502
b. Group insurance	503-509
2-a. Pension system	510-518
b. Incapacitated employees	519
c. Safety work	520
d. Medical service	520
e. Iron River Hospital	520-521
f. Physical examination of employees	521
g. Community health	522-523
h. Relief work	523
i. Employment	523
j. Improvement work	523
k. Community service work	523
1. Outdoor activities	524
m. The Mather Inn	524
n. Various department activities	524-525
o. Police Department	525
n. Appreciation	525
b. Whice and the	121
Perpert of Ceologist	and the second
Conergi	506 508
T Qtoff	500 522
I. DUALL	129-133 522 Flui
II. Geological and geophysical field work	733-744 FLF FF0
TV Curface am lengthing	545-550
IV. Surface exploration))1-)0(c(0 cp)
V. Underground exploration	500-5(4
VI. Land offers and outside explorations	272-279
VII. Microscopy	500-592
VIII. Other departmental highlights	593-595
	A CARLES AND A CARLES AND

Constant Trains

Sold Martine

RBM:PJB 10-25-57 -3-

Manager's Annual Report Year 1956

GROSS INDEX BY MINES

	Bunker Hill	Cambria- Jackson	Cliffs Shaft	Humboldt	Ohio	Maas	Morris
Ishpeming, Negaunee and Iron	CARL MARK		alterna i a se	all standings	1. 1. 1. 1.		
<u>River Districts:</u> General Production, shipments and	132-134	152	1	18	79	171-172	72
inventories Analysis Estimate of ore reserves Labor and wages Surface Underground or open pit	134-135 135-136 136-137 137-138 139	153-155 156 160 161-162 163	2-3 3-4 8-9 9-10 10	18-19 20 20-22 23	80-81 81 89 82-84 89	173-174 175 176-177 177-178 178-179	72-74 74 75 75 75 75-76
operations	139-145	164-167	11-16	23-29		179-183	76-78
Cost of opening, equipping, developing and operating Taxes Accidents & personal injuries Power New construction or equipment	146-149 150 150-151 151	157-159 168 169 170	4-8 16 16-17 17	30-40 41 41 42	85-88 90 90 90	184-186 187 187-188 188	
		Matl "A"	ner "B"	Pelletizing Plant	Republic	Spies	Tilden
Ishpeming, Negaunee and Iron River Districts: General Production, shipments & invento Analysis Estimate of ore reserves Labor and wages Surface Underground or open pit operati Cost of opening, equipping and developing Explorations Taxes Accidents & personal injuries New construction or equipment Power Nationality of employees	ries	A 43-44 45-47 48 56 57-58 59-61 62-68 49-55 69 70 71	189-190 191-193 194 199 200-201 202-203 204-213 195-198 214 215 216	217 218-219 220 221-222 241 242-243 223-240 244 245 245 246-247	91 92-93 94-96 97-98 98-103 104-114 117 118 118-121	254	122 123 124 126-128 124-125 129 125-126 130-131 128 128 128 128

Continued -

Manager's Annual Report Year 1956

GROSS INDEX BY MINES -2-

	Agnew	Alworth	Canisteo	Cushing	Hawkins
Mesaba District: General Production, shipments & inventories Analysis Estimate of ore reserves Labor and wages Surface Underground and open pit operations Beneficiations Maintenance and repair Cost of operation Exploration and future exploration Taxes Accidents and personal injury Proposed new construction Equipment received and proposed new equipment	255 256-257 257 257-258 258-259 259-260 260 260 260 260-261 262 262 262 262 262 262	263 263-264 264 265-266 266 267 267 267 267 268 268-269 269 269 269 269 269	270-271 271-273 273-276 276-277 278 279-282 282-284 284 285-287 287-288 288 289 290	291 291 292	293 294-295 295-296 297-298 299 299-301 301-308 308-310 310-311 310 310-311 311 312 312
	Holman- Cliffs	Hill- Trumbull	Sally	Open Pit Sargent	Wanless
Mesaba District: General Production, shipments & inventories Analysis Estimate of ore reserves Labor and wages Surface Open pit and underground Beneficiation Maintenance and repair Cost of operation Exploration and future exploration Taxes Accidents and personal injury Proposed new construction Equipment received and proposed new equipment	336 336-338 338-340 341-343 343 344 344-346 347-351 352 352-353 354 355 355 355 355	313-315 316-317 318-319 320-321 322 322-323 323-325 326-330 330 331-333 333-334 334 334 334-335	357 357 357-358 359 356 360 360 360 360 360 361 361 361 361	362-363 364 364-365 365-367 367-368 368 368-370 370 370 370-371 371 372 372 372 372 372	373 373-374 374-375 376 376 377-378 378 378 378-379 379-380 380 380 380

RBM:PJB 10-25-57 -3-

1

I. GENERAL:

The production in 1956 was 643,089 tons and the budget estimate was 510,600 tons. The above tonnage includes 14,256 tons of stockpile overrun. An operating schedule of 2-8 hour shifts per day with hoisting on 1-8 hour shift per day for 5 days per week was continued throughout the year. The mine was idle one day in April for installation of automatic cage hoist controls and 25 days in July and August due to the industrywide strike.

The cost of production was \$4.520 and the total cost was \$5.758 compared with \$5.852 and \$6.558 respectively in 1955. The lower costs in 1956 are attributable to increased efficiency with the new "C" Shaft plant as evidenced by the comparison of 7.36 tons per man per day in 1956 with 5.49 tons per man per day in 1955. The new plant made it possible to reduce the labor force by approximately 100 men and increased hoisting capacity speeded up all phases of the underground operations. A very favorable improvement was realized particularly in the tramming operation and output by the miners.

There was an average of 71 contracts working in the mine, the same as the previous year. Numerous mining areas were depleted, however better than budget production was maintained by re-entering old workings to recover relatively small areas and continuing a pillar recovery program. A few new mining areas were also brought into production.

Shipments totalled 656,485 tons leaving a balance of all grades of 24,723 tons on stockpile at the end of the year. This compares with 38,119 tons at the end of the previous year. The products were screened to -2" fines and -2" lump until May 20, 1956 at which time a screen-deck change was made and the products screened to -41" lump and -1" fines. The ratio of lump to fines was 75% and 25% with the 2" screening and a 82% and 18% ratio is being realized with the 1" screen deck. Second-class ore was processed through primary and secondary crushing and shipped as a third grade. This product averaged 15.6% of the total production.

Installation of the automatic pumping system under E. & A. CC-731 was completed and the pumps were placed in operation early in July. With the exception of necessary maintenance, all pumpmen labor has been eliminated with the installation of this system. Renovation of the old boiler house into a central shop building under E. & A. CC-730 was also completed. A reduction in the shop labor force and more efficient use of the shop labor has resulted from consolidation of the various scattered shops.

CLIFFS SH	LAFT MINE
ANNUAL	REPORT
YEAR	1956

II. PRODUCTION

a. Production by Grade and Months

2.20		CI	IFFS SHA	FT •		BANCROFT		S	ECTION 1	.0		
19 3 (A)	Optg.	111.1913	Cru	shed	-	Cr	ushed	7. 2. 6 2 11	Cru	shed	14. B 11.	
Month	Days	Lump	No. 1	No. 2	Lump	No. 1	No. 2	Lump	No. 1	No. 2	Total	Rock
Jan,	21	17,359	6,426	7,045	2,006	772	1,048	6,584	2,536	3,263	47,039	744
Feb.	21	21,592	8,357	8,102	1,900	724	1,434	6,589	2,508	6,047	57,253	1,460
March	22	22,018	8,640	8,319	4,274	1,651	1,399	5,386	2,067	3,122	56,876	1,536
April	20	18,471	8,866	8,557	5,383	2,584	647	3,910	1,876	2,170	52,464	1,440
May	23	31,902	10,534	3,067	5,268	1,739	192	12,623	4,169	1,200	70,694	1,136
June	21	32,588	5,436	10,229	3,683	614	1,159	11,435	1,908	2,862	69,914	954
July	0			St. Starting								
Auge	19	23,372	4,549	7,836	4,537	882	560	9,084	1,766	3,553	56,139	704
Sept.	19	22,632	5,405	5,796	3,986	952	120	8,806	2,103	2,021	51,821	252
Oct.	23	34,867	8,040	3,750	4,474	1,032	152	13,802	3,183	1,191	70,491	708
Nov.	21	32,848	7,332	1,625	4,844	1,082	70	15,203	3,394	309	66,707	424
Dec.	18	12,911	2,865	2,274	2,867	634	835	5,396	1,198	455	29,435	2,246
Total		270,560	76,450	66,600	43,222	12,666	7,616	98,818	26,708	26,193	628,833	11,604
Current	Yeart	S				T STANSA					28-34	
Stkp. 0	verrun	1,681	8,035		282	1,235		572	2,451		14,256	-
Total	228	272,241	84,485	66,600	43,504	13,901	7,616	99,390	29,159	26,193	643,089	11,604

b. Shipments

	Pocket Tons	Stockpile Tons	Total Tons 1956	Last Year Tons
Cliffs Shaft Lump	205,380	72,483	277,863	301,436
Cliffs Shaft Crushed #1	48,382	39,471	87,852	111,520
Cliffs Shaft Crushed #2	32,565	37,628	70,193	61,950
Bancroft Lump	32,001	12,372	44,373	43,316
Bancroft Crushed #1	8,419	6,057	14,476	12,655
Bancroft Crushed #2	2,293	4,342	6,635	9,922
Section 10 Lump	78,077	22,909	100,986	74,109
Section 10 Crushed #1	17,896	12,036	29,932	21,732
Section 10 Crushed #2	11,202	12,973	24,175	7,805
Total	436,215	220,271	656,485	644,445

c. Ore Statement

	On Hand <u>1-1-56</u>	Output For Year	Overruns	Total	Shipments	Balance on Hand
Cliffs Shaft Lump	9,374	270,560	1,681	281,615	277,863	3,752
C. S. Crushed #1	6,735	76,449	8,036	91,220	87,852	3,368
C. S. Crushed #2	11,337	66,600		77,937	70,193	7,744
Bancroft Lump	1,669	43,222	282	45,173	44,373	800
Bancroft Crushed #1	1,279	12,666	1,235	15,180	14,476	704
Bancroft Crushed #2	963	7,616		8,579	6,635	1,944
Section 10 Lump	3,184	98,818	572	102,574	100,986	1,588
Section 10 Crushed #1	2,208	26,708	2,451	31,367	29,932	1,435
Section 10 Crushed #2	1,370	26,193	-	27,563	24,175	3,388
Total	38,119	628,832	14,257	681,208	656,485	24,723
Total Last Year	131,254	512,705	38,605	682,564	644,445	38,119

2

2. PRODUCTION (Cont'd)

d. Working Schedule

The table below shows a comparison of working schedules for the past five years:

Year

Schedule

- 1956 2-8 hour shifts per day with hoisting on 1-8 hour shift per day, 5 days per week.
- 1955 3-8 hour shifts per day with hoisting on 2-8 hour shifts per day, 4 days per week Jan. 1st through April 15th - 5 days per week April 16th through November 28th - then 2-8 hour shifts per day with hoisting on 1-8 hour shift per day for the balance of the year.
- 1954 3-8 hour shifts per day with hoisting on 2-8 hour shifts per day, 5 days per week Jan. 3rd through March 31st - 4 days per week April 1st through December 31st.
- 1953 3-8 hour shifts per day with hoisting on 2-8 hour shifts per day, 5 days per week.
- 1952 3-8 hour shifts per day with hoisting on 2-8 hour shifts per day, 6 days per week May 1st to November 15th - 5 days per week November 15th through December 31st.

e. Production Delays

The following production delays were experienced during the year:

1. The mine was idle April 30th while automatic cage controls were installed.

2. The primary crusher motor burned out on March 15th. Repairs were completed so operations could be resumed on March 17th. Loss of production on this account was 3,900 tons.

3. Hoisting was discontinued after December 14th so that major repairs could be made to the supporting steel under the primary crusher. Repairs were completed so normal operations could be resumed on January 2nd, 1957.

3. ANALYSIS

a. Average Mine Analysis of 1956 Output:

	Dried	Phos.	Silica
Combined Cliffs Shaft Lump	59.86	.102	8.60
Combined Cliffs Shaft Crushed #1	56.03	.104	12.78
Combined Cliffs Shaft Crushed #2	53.30	.098	16.70

b. Average Analysis of Shipments for 1956:

	Grade	Iron	Phos.	Silica	Mang.	Alum.	Lime	Mag.	Sul.	Loss	Moisture
•)	Lump Ore	60.15	.104	8.40	.21	2.08	.83	.84	.005	.90	1.01
°)	Crushed Ore #1	56.40	.106	11.90	.21	2.54	.96	.99	.005	1.34	2.37
•)	Crushed Ore #2	53.60	.091	16.60	.26	2.10	.92	.91	.005	1.38	1.22

(°) Cliffs Shaft, Bancroft and Section 10 are combined.

3

3. ANALYSIS: (Cont'd)

c. Average Analysis of Ore in Stock Dec. 31, 1956

	and the second	Iron	Phos.	Silica	Mang.	Alum.	Lime	Mag.	Sul.	Loss	Moist.
Combined C. S. Lump	Dried	59.27	.109	9.94	.21	2.08	.83	.84	.005	.90	-
	Nat'1.	58.67	.108	9.84	•21	2.06	.82	.83	.005	.89	1.01
" C. S. Crush #1	Dried	54.70	.090	14.67	.22	2.54	.96	.99	.005	-	-
	Nat'l.	53.40	.088	14.32	.21	2.48	.94	.97	.005	1.31	2.37
" C. S. Crush #2	Dried	53.30	.084	17.07	.26	2.10	.92	.91	.008	1.38	-
	Nat'l.	52.65	.083	16.86	.26	2.07	.91	.90	.008	1.36	1.22

4. COST OF OPENING, EQUIPPING, DEVELOPING AND OPERATING

Comparative Mining Costs

and the second	1950	5	1955			
Product	643,0	089	551,	551,310		
Underground Costs	<u>Amount</u> \$2,244,236.49	<u>Cost/Ton</u> 3.490	\$2,363,487.97	<u>Cost/Ton</u> 4.287		
Surface Costs	303,575.40	.472	421,711.10	.765		
General Mine Expense	359,135.84	.558	441,264.22	.800		
Cost of Production	2,906,947.73	4.520	3,226,463.29	5.852		
Depreciation	532,613.18	.828	69,613.99	.126		
Taxes	200,391.00	.312	240,106.25	.436		
Loading & Shipping	63,238.01	.098	79,481.36	.144		
Total Cost at Mine	\$3,703,189.92	5.758	\$3,615,664.89	6.558		
Budget Cost of Production		-		6.219		
Number of Shifts and Hours	2-8 h	our	2-8 hou	ır		
Number of Days Operated	228		234			
Average Daily Product	2,821	fons	2,191 Tons			

5.46

344.58

2,016.04

16,054.91

IDLE EXPENSE ACCOUNT OF STRIKE Underground Costs Tramming Ventilation Pumping Compressors and Air Lines Underground Superintendence

27.78 Maint: Pockets and Chutes 1,348.31 " Mining Equipment 124.60 Telephones and Safety Devices 485.14 Holiday Pay 20,473.48 Total Underground Costs Surface Costs 1,286.26 Hoisting 847.91 Dry House 1,199.20 Policing General Surface 1,352.06 5.94 Maint: Other Mine Buildings Holiday Pay 244.26 4,935.63 Total Surface Costs General Mine Expenses Geological Department 778.01 2,125.91 Mining Engineering Department 1,199.99 Mechanical Engineering Department 612.00 Safety Department Research Laboratory 789.15 Analysis & Grading -- Laboratory " -- Shipping 842.87 440.00 Special Expense -- Pensions 9.00 " " -- Retirements " " -- Hygiene Clip 484.00 -- Hygiene Clinic 198.06 " - Employment Office 11 120.00 Ishpeming Office 8,220.00 Mine Office -- Supt. & Clerks 4,444.99 868.77 Central Warehouse Overhead Insurance -- Property 541.73 " -- Group, Health & Life " -- Group Annuity " -- Catastrophe 885.64 1,136.44 356.00 Personal Injury -- Comp. & Doctors 225.60 Taxes -- Unemployment Insurance " -- Old Age Benefit 63.43 335.82 1,722.03 Electrical Engineering Department Employees Insurance & Compensation 768.00 Total General Mine Expenses 27,167.44 Cost Of Production 52,576.55

5

COST OF OPENING, EQUIPPING, DEVELOPING & OPERATING: (Cont'd)

Detailed Cost Comparison

4.

	<u>Total 1956</u>		Total 1	.955	
	Amount	Per Ton	Amount	Per Ton	
Underground Costs					
Development	\$ 60,607.38	.094	\$ 160,212.82	.291	
Mining	1,129,590.70	1.758	1,066,731.39	1.935	
Tramming	286,410.72	.445	371,412.94	.674	
Ventilation	8,332.06	.013	11,691.51	.021	
Pumping	44.653.73	.069	42,998.25	.078	
Compressors & Air Lines	70.645.26	.110	104,934.93	.190	
Crushing & Screening - Undg.	77.147.51	.120	15.943.82	.029	
Underground Superintendence	159.668.98	.248	159.608.04	.290	
Maint: Pockets & Chutes	52.590.58	.082	66.424.70	.120	
Mining Fourpment	205.796.18	.320	207.454.97	.376	
Shaft	6.513.95	.010	10.136.04	.018	
Holiday Pay	27,620,92	-043	38,586,17	.070	
Vecetion Pay	85.502.01	-133	85.135.25	.155	
Telephones & Sefety	29,156.51	045	21,917.14	-040	
Matel Hadenersund Costs	0.011.006.10	2 100	2 262 107 07	1 207	
lotal Underground Costs	292449230049	3.490	2,303,401.71	4.201	
Surface Costs		110	06 000 00	7.56	
Hoisting	70,503.05	.110	80,233.28	•150	
Crushing & Screening - Surface	19,710.47	.030	80,522.48	.140	
Stocking	15,029.19	.024	115,014.71	•209	
Timber Yard	10,226.06	.016	-	10 8 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1	
Dry House	26,209.69	.041	32,013.25	.058	
Policing	13,721.14	.021	13,460.06	.024	
General Surface	18,810.83	•029	25,222.94	.046	
Maint: Headframe Bldg. & Equip.	11,537.81	.018	6,280.42	.011	
Other Mine Buildings	86,616.22	.135	26,745.45	.049	
Holiday Pay	5,919.19	.009	10,260.64	.019	
Vacation Pay	19,506.80	.031	21,358.81	.039	
Telephones & Safety	5,178.35	.008	3,255.09	.006	
Deferred Accts Top Tram Car	S. C. Sala	-	1,343.97	.002	
Total Surface Costs	303,575.40	.472	421,711.10	.765	
Geological Denartment	9.326.59	.015	13.545.62	.025	
Mining Engineering Department	19.588.77	.030	22,832,59	-0/1	
Mechanical Engineering Dept.	6.894.07	.010	15,111,88	.027	
Safety Denartment	6.688 20	.010	8,178,12	.015	
Bacarah Laboratorr	6 217 27	010	1. 062 71.	.009	
Analyzaia & Crading-Laboratory	25 050 62	010	43702.14	059	
analysis & Grading-Laboratory	1 067 01	.055	5 052 01	.010	
Special Emerge Densions	4,007.21	.000	121 22	.010	
Special Expense - Pensions	2 006 02	004	6 500 11	010	
" - Retirements	3,570.72	.000	0,5//.11	.012	
- Hygiene Clinic	4,793.12	.007	7,523.20	.014	
Tabaaning Office	1,025.02	.002	1,4/0,12	.003	
Isnpeming Office	73,911.15	•115	100,050.90	.181	
Mine Office-Supt. & Clerks	43,825.17	.070	40,091.27	.085	
Central Warehouse Overhead	14,301.55	.022	22,048.41	.040	
Insurance - Property	3,538.92	.006	4,285.92	.008	

6

COST OF OPENING, EQUIPPING, DEVELOPING & OPERATING: (Cont'd)

Detailed Cost Comparison (Cont'd)

		<u>Total 1956</u>			<u>Total 1955</u>		
General Mine Expenses (Cont'd)		Amount	Per Ton		Amount	Per Ton	
Insurance - Group, Health & Life " - Group Annuity " - Catastrophe Personal Injury - Comp. & Doctors	\$	32,885.85 12,256.10 4,537.58 15,500.29	.051 .019 .007 .024	\$	25,162.09 12,615.12 4,693.35 30,144.63	.046 .023 .009 .055	
Taxes - Unemployment Insurance Taxes - Old Age Benefit Electrical Engineering Department Design Department Employees Insurance & Comp.		9,592.76 31,850.00 10,996.19 1,834.92 6,903.72	.015 .050 .017 .003 .011		534.82 6,091.55 41,983.96 14,826.52 3,644.59 9,418.00	.011 .076 .027 .007 .017	
Power Credit Total General Mine Expenses		3,323.23 359,135.84	•005		441,264.22	.800	
COST OF PRODUCTION	\$2,	906,947.73	4.520	\$3	,226,463.29	5.852	

The lower cost, compared with 1955, is due chiefly to the higher production and increased efficiency realized with the new "C" Shaft plant. The wage increase, effective July 1st, and increased supply and material costs increased the expense in some accounts.

<u>Underground Costs</u> - The reduction in the labor force and increased efficiency brought about by the "C" Shaft plant accounts for the large reduction in underground costs. The largest cost improvement was in the tramming, mining and development accounts.

<u>Surface Costs</u> - The reduction in the surface labor force and the increased efficiency of the new surface flow-sheet accounts for the large decrease in costs compared with 1955. Repairs to mine buildings increased over last year.

<u>General Mine Expense</u> - The proportion of general expenses incurred at the Cliffs Shaft was reduced because of a major reduction in the labor force at the mine. Completion of the "C" Shaft plant eliminated the overhead expense on this project that was charged to operating.

Expenditure & Authorization Summary

E. & A. CC-705 - Gismo-Diesel Loader Transport

After a thorough trial of the Gismo Diesel loader underground, its performance was found to be unsatisfactory and the unit was returned to the manufacturer. A Diesel Traxcavator was then tested and after some modification its performance was found to be very satisfactory in the larger stopes. This machine is still undergoing further modifications to improve its performance. The project amount authorized was \$28,400.00 and the expenditure was \$16,879.76 in 1956.

E. & A. CC-730 - Automatic Pumping System

Work on the new pumping system was completed and the automatic pumps were placed in operation in July. The automatic pumps have lowered pumping costs as anticipated by eliminating the labor formerly charged to this account. The project amount authorized was \$109,770.00 and total expenditure was \$115,634.78. <u>CLIFFS SHAFT MINE</u> <u>ANNUAL REPORT</u> <u>YEAR 1956</u> 8

COST OF OPENING, EQUIPPING, DEVELOPING & OPERATING (Cont'd)

Expenditure & Authorization Summary (Cont'd)

4.

E. & A. CC-731 - Central Shop Facilities

The old boiler house building was converted into a Central Shops. This eliminated the various scattered shop buildings and reduced the shop labor force. Maintenance and repair of mine equipment is now accomplished with minimum handling and more efficient use of the shop crew. The project amount authorized was \$44,000.00 and the total expenditure was \$46,787.34.

E. & A. CC-793 - Dust Control System

Work on a dust collection system in the crusher station underground was started with installation of the equipment and duct work. This system is required to reduce the dust count in this room to an acceptable level. It was not possible to accomplish this using only fans and water sprays. The project amount authorized was \$27,850.00 and the expenditure totalled \$15,602.22 for 1956.

E. & A. CC-865 - Trolley Phones

This E. & A. covers the purchase of additional trolley phones for the underground locomotives. Trolley phones will be installed on all the motors on the lower levels where tramming has become more concentrated. This will provide greater protection against collisions and improve haulage efficiency. The amount of the authorization is \$4,300.00. There was no expenditure against this account in 1956. The phones are on order and delivery is expected early in 1957.

5. ESTIMATE & ANALYSIS OF ORE RESERVES

The reserves are estimated on the basis of the following factors:

High Grade of First Class Ore	-	8	cu.	ft.	per	ton	
Second Class Ore	-	9	cu.	ft.	per	ton	
Conglomerate & Second Class Ore	-	10	cu.	ft.	per	ton	

The annual increase in the estimated reserves is due largely to the method employed in making up the tax estimates. Experience has shown that extension of <u>proven</u> ore reserves for any considerable distance away from a working area has been an unreasonable assumption at this property due to the complex mine geology. Therefore, much of the proven ore reserves and additions are based on an accumulated group of more or less standard breast extensions. This practice accounts for nearly the same magnitude of proven ore reserves each year fluctuating somewhat according to depletion and ore development. No significant ore reserve additions were realized this year.

The following table shows a comparison of developed ore with the previous year as reported to the State Tax **So**mmission:

CLI	FFS	SH	IAFT	MINE
1	ANNUA	L	REP	ORT
28	YEA	R	195	6
	10-2010	100	100.00	Constant of the

ESTIMATE & ANALYSIS OF ORE RESERVES (Cont'd) 5.

	Cliffs Shaft	Bancroft	Section 10	Lease	Total Tens
Estimated Reserves-Dec. 31, 1955	666,653	88,165	346,866	435,031	1,101,684
Less 1956 Production	423,326	65,021	154,742	219,763	643,089
Balance as of 1955 Estimate	243,327	23,144	192,124	215,268	458,595
Estimated Reserves-Dec. 31, 1956	609,325	70,643	227,046	297,689	907,014
New Developed Ore	365,998	47,499	34,922	82,421	448,419

Expected Average Analysis of Ore Reserves

	Iron	Phos.	Silica	Mang.	Alum.	Lime	Mag.	Sul.	Loss	Moisture
Natural	57.50	.107	10.00	.35	2.10	.80	.80	.014	1.20	.85

The geological and engineering study of hard ore reserves, which was made in 1952, was reviewed in 1953, 1954, 1955 and 1956. Based on the 1956 revision and that portion of the second-class ore which will be mined, the proven and probable ore reserve is estimated at 2,403,992 tons as of December 31, 1956. The necessity of making a better #2 Grade, together with the additional information obtained from mining during 1956, accounts for the reduction in reserves as compared with 1955.

6. LABOR AND WAGES

Labor relations have been satisfactory although the Committee has been quite active in submitting complaints and a few formal grievances. There were 3 grievances submitted during the year Step of

Name	Nature of Grievance	Griev	ance Procedure
John Garrett, et. al. Charles Pelto John Smeltz, et. al.	Pay Shortage Pay Shortage Seniority	(3) (1) (3)	dropped settled settled
Employment			
No. of Men Beginning of Y	ear	400	
Separations	47		
Added During Year Decrease in Men	$\frac{7}{40}$	40	
Total End of Year		360	
Avg. No. of Men as per La (S	bor Statement tatistical)	378	
Avg. Absenteeism (S	tatistical)	26	

The following table shows a classification of separations in 1956:

	No. of Men
Quit	13
Retired	6
Transfers	19
Leave of Absence	2
Deceased	3
Absence for 2 years or more	
on account of sickness	4
Total	47

During 1956, 192 employees were entitled to 3-week vacations, 203 for two-week vacations and 23 for one week.

		YEAR 1956		
6. <u>LABO</u>	R AND WAG	GES (Cont'd)		and the second
	Statemer	nt of Wages		
	a.	Average Wages Per Day		
			<u>1956</u>	1955
		Total Surface & Underground	22.75	21.06
	b.	Average Wages Per Month		
			1956	1955
		Total Surface & Underground	472.06	410.67
		The mine operated on an average while averaging $19\frac{1}{2}$ days per month in	of 20 <u>2</u> days per mo 1955.	onth in 1956
	с.	Tons Per Man Per Day		
			<u>1956</u>	1.955
		Total Surface & Underground	7.36	5.49
Sall	d.	Labor Cost Per Ton		
		A STATE OF A	1956	1955

10

7. SURFACE

New Construction

The M.G. set for the Koepe cage hoist and all the necessary electrical controls was installed. Operation of the hoist on automatic control commenced the week of April 28th.

3.05

3.83

Total Surface & Underground

The old boiler house west of the engine house was converted into a central shops building in 1956. A small addition to the west end of this building was erected to house the bit shop. The renovation of the building was completed and the various shop facilities moved into this building in December.

Track was extended from the end of the tunnel to the west storage yard. This allows the direct movement of supplies from the shops or storage area to the shaft collar without any secondary handling.

The demolition of the old flow-sheet structures was completed and following this, tracks were re-located under the new pocket structure. The old shop building was converted into a garage for the mine mobile equipment.

A discharge line for the new pumping system was laid from the collar of "C" Shaft to a point several hundred feet northwest of the dryhouse. From this point, an open ditch was excavated to connect with the Carp River.

11

8. UNDERGROUND

A. General

Experience gained since the new flow-sheet went into operation, resulted in some revamping of dumping installations on the various levels at the orepass raises. Also, vibration caused damage to many structural supporting members and required replacement of the steel that was badly weakened by numerous cracks. Considerable re-enforcementswas also necessary to the supporting steel for the flow-sheet equipment.

The development program has continued on a small scale during the year due to lack of new areas to develop for mining. Continued stress on analysis, depletion of a number of stoping areas, ore tied up by haulage and the lack of possible ore exploration areas all contributed to a program of continual stoping area changes and revisions. Pillar recovery continued but at a higher rate than in 1955. The proportion of contracts on this work increased from 25% at the beginning of the year to 36% in December.

Increased pillar recovery has brought more contracts within the "A" Shaft and "A" Shaft East areas while the number of contracts in the Section 10 Lease area increased because of the development of an ore body extension. The number of contracts in the "B" Shaft, "A" Shaft Northeast and Bancroft Lease areas was reduced, however, due to depletion of ore areas.

The mine is logically split into mapping areas in which "A" and "B" Shaft map units represent the inlying areas in which a large proportion of the contracts are mining in old stopes. Bancroft and "A" Shaft Northeast, "A" Shaft East and the Section 10 Lease and Moro Mine represent the outlying or fringe areas which must be depleted prior to the inlying areas for orderly mining.

The New York Mine, also known as the Harlow Clark Estate, occupies the SEL of the SEL of Section 3 at the extreme northeast corner of the Cliffs Shaft Mine. An offer to lease this property was received in the latter part of 1956 but after an investigation and evaluation the option to lease was turned down. The very small reserves that are indicated and the considerable rock development that would be necessary rules out the possibility of an economical mining proposition in this property.

B. Mining Area

1. <u>"A" Shaft East</u> - (East of 2800 E and extending from the south boundary of the Bancroft Lease to 1200 S)

No development was done in the "A" Shaft East area this year. All known ore areas are presently being mined.

The number of contracts here was increased to enable a pillar recovery program to be started in addition to the other mining.

The remaining reserves here lie between the 4th and 8th level. Emphasis on mining the fringe areas will continue to be stressed as we attempt to deplete the outlying ore structures before retreating towards shaft.

8. UNDERGROUND (Cont'd)

B. Mining Area (Cont'd)

2. "B" Shaft - (West of 400E)

Development in "B" Shaft consisted of driving sub-level drifts and short raises to develop ore body extensions and provide travelling roads. One raise was driven from the 5th level Barnum to the 1145' sub-level to develop ore encountered in U. H. No. 805. A branch was also driven from this raise to a stope on 1st level "B" Shaft to facilitate mining out the bottom in the stope.

The number of contracts working in "B" Shaft was reduced to 13 in 1956. Seven of the contracts are mining on or above 1st level and 6 have only pillars as reserves.

The bulk of the first class ore reserves continue to lie above 1st level. The "Lake Bancroft" structure, drilled in 1953, continues to be mined below 2nd level elevation and appears to get narrower as mining progresses downward.

3. "A" Shaft - (400 E - 2800 E, between the Bancroft & Section 10 Leases)

Some main-level drift was advanced on 8th level to by-pass a section of the old "main line" haulage drift and thus make available for mining a large floor pillar. One raise and two branches from this main raise were driven from 15th to 10th level in the vicinity of old #46 Contract stope to develop ore encountered in several drill holes. Several contracts advanced sub-level drifts and short raises in their respective stopes to develop ore body extensions and provide travelling roads.

Nine contracts have no reserves other than upright and floor pillars. Reserves in "A" Shaft are dispersed and in part, tied up by tramming operations. During the year, mining continued on the 6th level floor pillar and a section of the 8th level floor pillar was also made available for mining. Other such areas will be opened to mining after relocation of travelling and tramming roads.

4. Bancroft and "A" Shaft Northeast

One raise was driven from 10th to 9th level in the "A" Shaft Northeast area to facilitate mining in a stope on 9th level and to make a floor pillar available for mining. Several contracts advanced sub-level drifts and short raises in their respective stopes to develop ore body extensions and provide travelling roads.

The number of stoping contracts remained the same as in 1955 although there was some shifting of contracts in the various stopes. Reserves of ore in the Bancroft Lease and the "A" Shaft Northeast area are now concentrated between the 9th and 11th levels although a small tonnage remains at the 5th level elevation. Three of the contracts in this area have only pillars as ore reserves.

5. Section 10 Lease and the Moro Mine

Development in the Section 10 Lease was fairly extensive during the year. One raise was driven from 8th to 5th level to develop an ore structure adjacent to #80 Contract shrinkage stope. Considerable sub-level drifting was done above 9th level elevation to outline the ore body and develop a fairly large area for mining. Mining, during the year, ranged in elevation from the 5th to 10th level with most crews concentrated between the 5th and 8th levels. Two contracts were added to this area in 1956 and with the present concentration of contracts this is now the most productive area in the mine.

8. UNDERGROUND (Cont'd)

C. Delimiting Ore

The underground diamond drilling program was completed in 1955. There was no drilling done underground or on surface for hard ore reserves in 1956.

A drilling program has been continued from surface to explore for deeplying soft ore reserves that would be mineable from the "C" Shaft. This exploration is summarized in the following paragraphs.

<u>Section 10, 47-27 - Deep Exploration</u> - Drilling in Section 10, 47-27 has continued throughout 1956 in one hole, D.D.H. No. 29. This hole was located on the basis of the geology predicted in the Deep Soft Ore Study.

D.D.H. No. 29 is located on the old Lake Mine surface on the north shore of Lake Angeline in the NWL of the SEL of Section 10, 47-27. The hole, which was drilled to a depth of 855' by the end of 1955, was continued to a depth of 4950' by the end of 1956. Intrusive and unoxidized iron-formation have been penetrated and drilling will be continued until the footwall Siamo slates are encountered. Operations at this hole have been hampered by a series of accidents consisting of the loss of strings of drill rods down the hole twice during the year and the collapse of some BX casing. The BX casing was placed in the hole to prevent the possible disturbance of any of the 5 Hall-Row wedges in the hole. Total down time while rods were theing recovered, etc. amounted to 110 days.

Section 3, 47-27 - Deep Exploration - Drilling in Section 4, 47-27 was resumed to confirm the knowledge previously gained by drilling and to expand and further outline the predicted ore body. Drilling commenced on the first hole in March and on the second hole in August. Both holes were planned and located on the basis of the geology interpreted from previous drill hole information and that predicted in the Deep Soft Ore Study. The first hole, D.D.H. No. 52, was located approximately 150° due north of D.D.H. No. 44 and was drilled to confirm the cross section interpreted from D.D.H.s No. 43 and No. 44. The second hole is located midway between D.D.H.s No. 37 and No. 44 in an effort to prove or disprove a continuous ore structure between these holes.

D. New Equipment

A large amount of the equipment purchased in 1956 was related to the E. & A. Projects. However, a substantial amount was required for maintenance of existing plant and replacement. Miscellaneous spare parts for the new flow-sheet were also purchased and placed in stock. The following major items were purchased in 1956:

Item	Amount
33 - Fire Extinguishers	\$ 1,744.77
1 - Jackabi Tachoscope	135.73
1 - 60 ^t Aluminum Ladder	445.50
1 - Exide Locomotive Storage Battery	1,931.75
6 - Kellog Phones	366.26
2 - Vertical Hot Water Tanks & Heaters	952.38
4 - T-9 Hand Trucks (Welding Tanks)	116.00
6 - 25 H.P. Scraper Hoist Motors	3,642.00
1 - #448 Valve Body Assembly	590.55
8 - Roller Smith Galvanometers	326.01

8. UNDERGROUND (Cont'd)

D. <u>New Equipment</u> (Cont'd)

6 - DuPont CD-2 Blasting Machines \$ 2 - Karbide King Saws 1 - Hand-Type Machine Cleaner 1 - Intercooler 3 - Heat Regulators 12 - Descalers	831.85 177.30 142.60 723.88 405.00 231.60 1.474.00
2 - Karbide King Saws 1 - Hand-Type Machine Cleaner 1 - Intercooler 3 - Heat Regulators 12 - Descalers	177.30 142.60 723.88 405.00 231.60 1.474.00
1 - Hand-Type Machine Cleaner 1 - Intercooler 3 - Heat Regulators 12 - Descalers	142.60 723.88 405.00 231.60 L.474.00
1 - Intercooler 3 - Heat Regulators 12 - Descalers	723.88 405.00 231.60 L.474.00
3 - Heat Regulators 12 - Descalers	405.00 231.60 L.474.00
12 - Descalers	231.60
10	1.474.00
12 - Unit Heaters	
6 - Valves	1,099.10
1 - Fig. 12 Expansion Joint	1,165.97
1 - EJ-2 Thor 3/8" Drill	48.75
1 - 5V Thor Air Grinder	262.45
1 - #80 Thor Impact Wrench	435.00
1 - Electric Hoist (H-5 - 5 Ton Lo-Hed)	1,828.00
5 - Mule Hoists (No. 101 - 12 Ton)	188.15
50 - Pacific Clevis Blocks	2,828.25
1 - Steel Scaffold	1,950.38
1 - Sump Pump	121.37
1 - Set of Track Links for Traxcavator	634.02
1 - Lincoln Welding Machine	632.50
1 - Electric Starter (Westinghouse)	1,403.52
4 - Westinghouse Disconnect Switches	1.62.00
1 - Industrial Fork Lift Engine	477.00
1 - Pump & Injectors (Euclid)	553.17
Miscellaneous Spare Parts for	
Undg. Flow-Sheet <u>2</u>	9.885.92
Total \$5'	7,912.73

E. Explosives

Powder consumption decreased considerably compared with 1955 due, principally, to a large carry-over of broken ore in the stopes from 1955. Discontinuance of the use of 2" powder in secondary blasting also effected some reduction in consumption. The average price of powder increased from \$17.50 per cwt. to \$18.18 per cwt.

Starting in January, DuPont explosives replaced Hercules explosives throughout the mine. In April, 1956, a credit of \$5,370.40 was received from DuPont in payment for defective powder supplied to the mine in the first quarter of 1956. The net result of the increase of explosive cost, the credit for defective powder and the discontinuance of the use of 2" powder in secondary blasting, was a decreased explosives cost per ton which is summarized in the following tables:

TABLE I

Cost of Explosives - Operating

and the second second and the second	Quantity	Avg. Price	1956	1955
Powder, Lbs All Kinds Misc. Supplies (Cans. Fuse. Testers.	513,300 etc.)	18.18	\$ 87,941.12 51.149.22	\$111,590.55
Total			\$139,090.34	\$158,913.17

14

TABLE II

8. UNDERGROUND (Cont'd)

E. Explosives (Cont'd)

Unit Cost and Consumption	of Explosives	
	<u>1956</u>	1955
Pounds of Powder Per Ton of Ore Tons Ore Per Pound of Powder	•798 1.253	1.157 .865
Cost Per Ton For Powder Cost Per Ton For Blasting Supplies Cost Per Ton For All Explosives	•137 •080 •216	•202 •086 •288
TABLE III		
and the second second second second second second	<u>1956</u>	1955
Cost Per Ton Developing	.001	.046
Cost Per Ton Mining	.215	.239
Total	.216	-285

F. Tungsten Carbide Insert Bits

Since tungsten carbide insert bits were introduced into Cliffs Shaft, a footage record per bit has been kept for the purpose of determining the performance of various type bits and a comparison of costs. Numerous bit tests were made during the year and the results show the Ingersol-Rand Carset bit is still superior to other types. The following table shows the pertinent data:

	Descrip	otion	Quantity	Price	1956	1955
Ing. Rand,	Series	113 - 1-3/8"	3,264	11.00	\$35,904.00	\$27,213.12
H	Ħ	113 - 1-1/2"				2,262.00
11		113 - 1-5/8"	50	13.65	682.50	322.50
H		115 - 1-5/8"	280	12.90	3,612.00	19,143.60
		115 - 1-3/4"	50	14.73	736.25	884.40
11		115 - 2"				158.95
Rockbits	H	113 - 1-3/8"	125	10.30	1,287.50	3,038.50
1		115 - 1-5/8"				524.40
n	11	115 - 2-1/8"	24	19.15	459.60	-
H	11	HM = 2-1/4"			•	57.30
Copco		113 - 1-3/8"			- 11 - 12 - 12 - 13 - 13 - 13 - 13 - 13	784.00
Vascoloy		113 - 1-3/8"			and the second	475.00
Kennametal		113 - 1-3/8"	31	10.90	337.90	-
	Tot	tal	3,824	11.25	\$43,019.75	\$55,020.85
	Product	tion - Tons			643,089	551,310
Cost Per Ton of Ore Produced Feet Drilled - Rock & Ore		d		.067	.100	
				628,337	867,271	
	Average	e Feet Drilled Per Bi	t		164	183
	Cost P	er Foot of Hole			.068	.063
					the second se	

15

16

8. UNDERGROUND: (Cont'd)

F. Tungsten Carbide Insert Bits

The average price per bit decreased slightly compared with 1955. This decrease was effected by the continued conversion of mining contracts from the 1-5/8" bit to the smaller 1-3/8". The average footage per bit decreased because the greatest amount of footage was drilled in ore and very little was drilled in rock. The lower footage per bit is also partly due to the increased production of the normally very hard secondary grade ores. The bit cost per foot of hole is nearly the same as in the previous year.

G. Pumping

The development of the sump and installation of the automatic pumps was completed and the plant was placed in operation in July. Two 500 g.p.m. centrifugal pumps, that were received from the Spies Mine, are connected in parallel and operate as a unit. A 1375 g.p.m. centrifugal pump is used as a standby. This system has been operating very satisfactorily.

9. TAXES

Comparative data for 1956 and 1955 is shown below:

24 Children Charles and State	<u>1956</u>		1955	
The second s	Valuation	Taxes	Valuation	Taxes
Realty	3,600,000	146,880.00	3,200,000	131,200.00
Personal	451,100	18,404.88	1,116,100	45,760.10
Lot 2, Sec. 3,47-27, Bancroft	545,000	22,236.00	580,000	23,780.00
Lot 174, Nelson's Addition	100	4.08	100	4.10
S. 35.91' of Lot 179	50	2.04	50	2.05
St of NWL of Section 10,47-27	900,000	36,720.00	960,000	39,360.00
Total Cliffs Shaft Mine	5,496,250	224,247.00	5,856,250	240,106.25
Taxes Per Ton Produced		•349	C. States	•436
Taxes Per Ton Shipped		.348		•373

10. ACCIDENTS & PERSONAL INJURY

Nine compensable and fourteen non-compensable injuries occurred in 1956 for a total time lost of 491 days. This record of 23 accidents, compared with the 37 incurred during 1955, ranked Cliffs Shaft second on safety during 1956 among the Company's underground mines.

Comparison of Frequency and Severity in 1956 and 1955 is as follows:

Year	Frequency	Severity
1955	20.04	.771
1956	33.65	.718

Frequency Rate - Number of accidents for every 1,000,000 man hours Severity Rate - Number of days lost per 1,000 man hours

14

10. ACCIDENTS & PERSONAL INJURY (Cont'd)

A summary of the compensable accidents is listed below:

#1349 - <u>Nels Wirtala</u> - Scraper Operator - January 23, 1956 - slipped and fell while carrying a block and injured right wrist - lost time, 65 days.

#1350 - <u>Clayton Weinberg</u> - Contract Miner - March 17, 1956 - struck on left arm by rope socket and clevis which broke from a scraper - fracture af the upper and lower left arm - lost time, 150 days.

#1351 - Oscar Stolen - Motorman - August 7, 1956 - left foot and ankle - slipped between box and underground car while dumping car. Contusions, no fracture - lost time, 37 days.

#1352 - William J. Johnson - Motorman - August 23, 1956 - slipped and twisted neck while avoiding chunk coming through chute - time lost, 8 days.

#1353 - John Armitage - Contract Miner - September 7, 1956 - struck on right foot by chunk while barring. Severe contusions and abrasions. Lost time, 62 days.

#1354 - Albert Manty - Car Repairman - September 26, 1956 - strained back while lifting air tugger. Low back pain - time lost, 7 days.

#1355 - <u>Alfred Hendra</u> - Contract Miner - October 16, 1956 - strained right arm while lifting drill machine. Rupture of tendon of right bicep muscle-time lost, 56 days.

#1356 - John Bess - Contract Miner - December 14, 1956 - Hit on right elbow by jackleg when machine fell. Broken bone right arm. Time lost - 8 days in December, and 40 more days estimated in 1957.

#1357 - Francis Marketty - Motor Brakeman - December 20, 1956 - left foot squeezed between drawhead and car when he stepped between train of cars. Compression fracture, left medial cuniform. Time lost, 4 days in December and 20 estimated days in 1957.

11. POWER

Total Cost	<u>1956</u> \$81,464.60	<u>1955</u> \$97,017.06
K. W. H.	8,960,441	10,412,790
Average Cost Per K.W.H.	.0090916	.0093171
K.W.H. Per Ton	13.9	18.9
Cost Per Ton	.127	.176

The decrease in K.W.H. consumption per ton is due to discontinuing operations in the old "A" and "B" Shafts on a two-shift basis and replacing them with the "C" Shaft operation on a one-shift basis. Power consumption for the air compressors was also reduced with the one-shift hoisting schedule in "C" Shaft. Lower power consumption and lower peak demand, which is characteristic of the Koepe hoist, compared with a drum hoist, is reflected in the lower energy cost.

HUMBOLDT MINE ANNUAL REPORT 1956 YEAR

1. GENERAL:

The Humboldt Mine discontinued the production of concentrate at the end of October, 1955, and remained idle until April 18, 1956. During the first two months of the year, there was limited activity at the mine with repairs to the pit equipment being resumed by late February and repairs in the mill and crushing plants beginning in March. 18

Following the startup, the mill ran $7\frac{1}{4}$ months producing 206,901 tons of concentrate for an average of 28,500 tons per month. This can be compared to the 1955 monthly average of 21,500 tons. Also, during the year, a total of 519,904 tons of crude ore, 47,026 tons of waste, 191,545 cubic yards of earth and 64,697 cubic yards of rock were handled at the Humboldt Mine. The grade of the concentrate produced was 61.80% iron, 0.084% phosphorous, 9.50% silica, 0.018% sulfur and 6.64% moisture which is practically identical to the guarantee.

During the operating months, the mill produced concentrate on a schedule of twenty shifts per week while seven day shifts per week were scheduled for mining ore. Thirteen additional shifts (6 days-7 afternoons) were utilized for stripping and waste removal throughout the summer. Because of the heavy shipping schedule and the need for moving a large yardage of waste rock associated with the ore, the development of the pit to the north was retarded slightly during the year. For this reason, an additional pit crew for stripping overburden began work in November adding five midnight shifts to the schedule described above.

The steel strike terminated operations at the end of June and the mine did not resume production until August 10th. This date was four days following the general settlement as repairs to the grinding section of the mill and the primary crusher necessitated the additional idle time.

A monthly production record was set during June when 34,156 tons of concentrate were produced.

At the end of the year, the pit was developed throughout the areas known as knobs one, two, three and four. This total pit length encompasses an area bordering near the primary crushing plant to the south and extending northward for 2,500 feet.

2. PRODUCTION, SHIPMENTS AND INVENTORIES:

L.	Operating Schedule:	No. of Days	Shifts Per Day	Hours Per Shift	Total <u>Shifts</u>
	Mining	210	1 & 2	8	287
	Hauling and Crushing	210	1 & 2	8	287
	Mill Operating	210	1 & 2	8	595

ELCONALIN

HUMBOLI	DT MINE
ANNUAL	REPORT
YEAR	1956

1:

2. PRODUCTION, SHIPMENTS AND INVENTORIES: (Con't.)

b. Production by Months:

and the second second second	Tons o	f Crude	Tons of Concentrate
Month	Crushed	Milled	
April	26,948	28,233	9,749
May	78,551	78,394	28,063
June	79,660	78,005	34,156
July	a production of the second	-	
August	52,327	51,898	19,227
September	75,109	73,654	30,795
October	72,264	71,796	29,636
November	70,346	70,806	29,478
December	64,699	65,258	24,000
Stockpile Overrun	1. 1. <u>.</u>	States a	1.797
Total	519,904	518,044	206,901

c. Production Averages:

1956	1955
2,476	2,009
985	745
33.06	30.30
13.16	11.05
39.94	36.51
	<u>1956</u> 2,476 985 33.06 13.16 39.94

 Shipments	Tons	Iron	Phos.	Sil.	Sul.	Moist.
On hand - December 31, 1955	56,596	61.60	.094	9.43	.009	7.84
Production - 1956	206,901	61.80	.084	9.50	.018	6.64
Stockpile to Presque Isle	122,202	61.29	.088	9.90	.011	6.88
Pocket to Presque Isle	90,149	62.01	.088	9.19	.020	6.47
*Pocket to O. Bohlin & Son	134	61.50	.084	9.01	.288	6.57
Stockpile to M. A. Hanna Co.	2	62.25		9.88		
On hand - December 31, 1956	51,010	61.65	.077	10.01	.013	6.84
	States (C. S.	the ground of the local states		F 12 - 12 F 7 - 5 P	1. Control 2, 20 (1997)	a the second second

* All Rail

Ch:

Estimated Production and Analysis - 1956:

and Analysi

	Tons	Iron	Phos.	<u>Sil.</u>	Sul.	Moist.
Dried		61.80	.090	9.50	.010	
Natural	179,985	57.17	.083	8.79	.009	7.50

HUMBOLDT	MINE
ANNUAL R	EPORT
YEAR -	1956

COLLON-DAK

CIFEEKL EOMO

21

3. ESTIMATE OF ORE RESERVES:

a. Estimated Reserves:

1956 Estimate - 40% Recovery

Total Crude	Humboldt Mining Co.	Weber	Total
30,063,600	11,204,240	821,200	12,025,440
7,566,100	2,776,300	250,100	3,026,400
7,514,600	2,758,000	247,800	3,005,800
45,144,300	16,738,540 1	,319,100	18,057,640
	30,063,600 7,566,100 7,514,600 45,144,300	International Crude Humboldt Mining Co. 30,063,600 11,204,240 7,566,100 2,776,300 7,514,600 2,758,000 45,144,300 16,738,540	Total CrudeHumboldt Mining Co.Weber30,063,60011,204,240821,2007,566,1002,776,300250,1007,514,6002,758,000247,80045,144,30016,738,5401,319,100

The State State	<u>S</u>	tripping	
	Cu. Yds.	Cu. Yds.	Cu. Yds.
	Surface	Rock	Pit Waste
1210 ¹ Elevation	1,712,455	8,826,603	1,701,742
1110 ¹ Elevation	503,800	6,040,700	316,395
TOTO. FIEVation	575,200	7,790,500	308,800

4. LABOR AND WAGES:

a. General:

The hourly rate crew at Humboldt averaged 95 men from April through October and 110 men during November and December. This increase reflected the addition of a stripping crew, the need for extra men in the crushing plants during the winter season and a slight increase in the maintenance gangs.

In order to increase the supervisory staff at Humboldt, hourly rate leaders were utilized for operating the crushing plants and heading the mill maintenance crews.

Labor relations continued on a high level as no formal grievances were filed. However, three men were discharged during the year because of absenteeism and stealing.

The following changes were made involving salaried personnel during 1956:

Name	Title	Status	Date
Lester Lundin	Master Mechanic	Placed on Salary	April
Otto Urpila	Mill Foreman	Placed on Salary	April
Vivian Delbridge	Chemist	Placed on Salary	April
Harriette Dougherty	Stenographer	Hired	April
James Villar	Geologist	Hired	July
James Villar	Geologist	Hired	Jul

RIJJ BROUND

HUMBOLDT MINE ANNUAL REPORT YEAR 1956

4. LABOR AND WAGES: (Con't.)

b. Report of Men Hired, Transferred and Separated:

Month	First of <u>Month</u>	Hires	Trans. from Other Mines	Separated	Trans. To Other Mines	End of Month
January	1			and the second		1
February	1	4	A CARLER AND			5
March	5	23	a state of the second second	Contraction and		28
April	28	62	2	Contraction Booking		92
May	92	6	2		1	99
June	99		1	Section 18	3	97
July	97					97
August	97	1	1	2		97
September	97	3	1		and the second	101
October	101	ì	2			104
November	104	4	8	2		114
December	114		2	3	The sea were	113

The above table includes only hourly and salaried men on the Humboldt Mine payroll.

c. Report of Vacations Paid:

OR COL.

Year	No. Men	Total Hours	Total Amount	Avg. Rate Per Hour
1956 1955	90 95	5,360 5,000	\$ 12,965.28 \$ 11,405.00	2.419 2.281
IL BYON	STR.			Carl Carl St

-4-

HUMBOLDT MINE ANNUAL REPORT YEAR 1956

2:

4. LABOR AND WAGES: (Con't.)

e.

d. Annual Statement of Labor:

Hourly Employees	Sta. Men	Hours	Amount	Avg. Rate
Straight Time Overtime Shift Differential - Afternoon Shift Differential - Night	49출	$130,678\frac{1}{2}$ 4,673 36,459 13,308	\$300,347.04 5,629.90 2,220.01 1,231.69	2.298 1.205 .061 .092
Holiday Worked		2,472 789	1,841.20	2.333
Premium Time Total Hourly Employees	493	5,860 ¹ / ₂	<u>1,527.13</u> \$330.102.02	.260
Salaried Employees				
Mine Payroll Total Mine Payroll	2 ¹ / ₂ 52 ¹ / ₄	6,538 ¹ / ₄ 137,216 ¹ / ₄	\$ 21,213.94 \$351,315.96	3.244 2.560
General Payroll	Col	1.Owne	(CBC	
Salaried Straight Time Overtime	3	7,793	\$ 20,525.62 332.83	2.634
Total Labor	60 ¹ / ₂	158,5354	\$413,983.56	2.611
Distributed as Follows:				
Idle Expense Operating Humboldt Mine Winter & Idle Expense Stripping Uncompleted Construction	3年 48 3章 6 1十	2,199 125,805 ¹ / ₂ 8,890 ¹ / ₄ 15,525- 3,460	\$ 5,865.64 326,529.60 22,446.53 40,634.34	2.667 2.595 2.525 2.617 3.083
Other Mines		1,338	4,053.81	3.029
Total as Above	60 ¹ / ₂	158,5354	\$413,983.56	2.611
Labor Cost:	as - Tong T	lone	206 901	
Number of Days Operated Number of Shifts Operated Average Daily Production	200,901 210 595 985			
Average Production Per S Tons of Concentrate Per 1 Average Wages Per Man Day Average Job Class	348 13.16 \$20.76 10			
Total Amount Paid for La Labor Cost Per Ton of Con	\$326,529.60 \$1.578			



5. GENERAL SURFACE:

a. Buildings and Repairs:

A small concrete extension along the west dumping area of the primary crusher building was added during the fall. This extension formed a box which permitted the slide dumping of crude ore to the crusher.

To improve the storage facilities at the mine, the principal warehouse within the shop building was completely enclosed by placing a roof over the area. In order to facilitate the storage of additional electrical supplies, a small wood framed warehouse was erected immediately west of the shop-garage building.

A new 20,000 gallon fuel tank was installed in an excavation east of the garageshop building to handle a heavier furnace oil and to allow for storage of larger quantities of fuel. At the end of the year, only the plumbing work remained to be completed on this project.

b. Water Supply:

Lake Lory continued to supply fresh water make-up for the Humboldt plant. The elevation of water in the lake ranged from a high of 1560.1' in May to a low of 1558.0' in October. Periodic checks were made of the solids content of the tailing dam overflow entering Lake Lory. The amount of solids entering the lake in parts per million was about twenty with highs of forty-five being reached during periods of intense or lengthly rainfall. Such action increased the water flow rate through the swamp and hindered the normal settling and filtering action. This situation was also true during the winter months when the vegetation was dormant and much less filtering medium was present.

c. Roads, Transmission Lines, Etc:

The transmission line from the main breaker station, feeding power to the pit, was replaced in March with heavier copper wire. This new line, running 1,600 feet in length, was necessary to avoid voltage drops hampering the pit operation.

6. OPEN PIT:

a. Stripping:

The surface overburden stripping program began in April and continued throughout December except during the steel strike in July. The bulk of this stripping was for the opening of the 1545 foot bench along the northern end of the pit. Some dragline work was completed on the footwall side of the third knob. After November 15th, all dragline work was abandoned due to freezing conditions and only shovel work was undertaken after that date.

The rock stripping during 1956 represented material from both the first and third knob areas. This material was made up of dikes and associated unoxidized iron formation that intruded the ore body. Also partially completed in the first knob area was a footwall access road to the primary crusher extending from the plant down to the 1585 foot bench.

HUMBOLDT MINE ANNUAL REPORT YEAR 1956

6. OPEN PIT: (Contt.)

a. Stripping:

The following table summarizes the stripping for 1956:

Month	Surface	Rock	Total	Man Davs	Man Dav
January	and the second	Salar Salar	a la serie de la s	1997年1月1日日	and a start hard
February					
March				26	
April	7,745	3,804	11,549	122	94.66
May	4,355	9,930	14,285	133	107.41
June	35,480	5,596	41,076	296	138.77
July	FIGLIFIN			-	and states
August	20,620	7,840	28,460	166	128.36
September	33,895	14,043	47,938	213	225.06
October	51,975	8,520	60,495	325	186.14
November	14,635	8,580	23,215	256	90.33
December	22,840	6,384	29,224	403	72.52
Total - 1956	191,545	64,697	256,242	1,940	132.08
Total - 1955	227,670	154,588	382,258	3,955	96.65
Total to Date	719,706	255,695	975,401	9,2731	105.18

b. Stripping Expenditures:

	Amount	Amount	Amount
	Authorized	Expended	Unexpended
E&A HM - 12	95,400.00	95,400.00	4,264.52
E&A HM - 22	151,182.00	146,917.48	

-7-

HUMBOLDT MINE ANNUAL REPORT YEAR 1956

6. OPEN PIT: (Con't.)

c. Open Pit Mining:

A total of 514,844 tons of crude ore was hauled to the crusher from the pit. The majority of this tonnage was mined from the 1545 foot bench in the first knob area with the remainder from the 1580 foot bench of the third knob area. The crude ore from the first knob was of good grade except for those areas associated with igneous dikes. Unoxidized and lean iron formations were experienced along these intrusives making selective mining difficult. This resulted in a lowering of the grade of crude ore delivered to the crusher. The material from the third knob area was also lean with the analysis for all crude mined for the year averaging 29.7% Fe.

Most of the primary drilling was accomplished by the jet piercing machine. Wagon drills were used to drill small areas in difficult terrain and where sloping banks needed relief for the jet pierced holes. To realize the savings from deep, widely-spaced jet holes, 45 foot holes at spacings from 21' x 21' to 26' x 26' were utilized for blasting throughout most of the year.

A new $5\frac{1}{2}$ inch Gardner-Denver wagon drill was under test during the fall. The exceptionally good bit footages realized during the test indicated reasonable costs for drilling with this machine. Four and one half inch diameter bits were tested in both the crude ore and hanging wall conglomerate.

In addition to the two regular Humboldt shovels, an additional 54-B was transferred to the Humboldt Mine during the fall. This shovel greatly aided pit development as it not only served as a spare shovel if a breakdown occurred, but reduced the long moves required as the mining faces and stripping areas became widely separated.

There were two major breakdowns that hampered the pit operation involving the P&H shovel. In November, the spur gear driving the left track broke, while in December, the boom developed cracks that needed considerable welding.

As the 1545 foot bench was opened during the year, large quantities of water were encountered resulting in the pumping of up to 800 gallons per minute to maintain a dry pit.

-8-

HUMBOLI	OT MINE
ANNUAL	REPORT
YEAR	1956

6. OPEN PIT: (Con't.)

3 ER

16:001

c. Open Pit Mining:

Summary of Pit Production

Crude Ore from Pit to Crushing Plant	and the second second second second	514,844	tons
Crude Ore from Stockpile to Crushing H	lant	5,060	tons
Total Crude Ore from Pit to Crushing F	lant	519,904	tons
Crude in Stockpile - December 31, 1955	;	36,597	tons
Crude Ore from Pit to Stockpile - 1956		ACTION OF A	1000
Crude Ore from Stockpile to Crushing H	lant - 1956	5.060	tons
Crude Ore in Stockpile - December 31,	1956	31,537	tons
Waste Rock, Pit to Dump - Yards		23,513	yards
Total Footage Drilled Jet Piercer	States and a state of the second	20,470	feet
Total Footage Drilled Wagon Drills		24,320	feet
Jackhammer Hours - Secondary Drilling		394	hours
Average Grade of Crude Ore	29.70 Iron	56.45%	Silica
Cost Per Ton of Crude Ore		.783	
		the second se	

Working Schedule

Wagon Drilling			2	Shifts	Per	Day	-	5	Days	Per	Week
Jet Drilling	2	&	3	Shifts	Per	Day	-	5	Days	Per	Week
Crude Ore Hauling	1	&	2	Shifts	Per	Day	-	7	Days	Per	Week

Summary of Powder Used:

Primary Blasting - Jet Holes

Type	Unit Cost	Amount Used	Total Cost
7/8"x 8"Gelamite	19.90 C	50 lbs.	\$ 9.95
2"x 8"Gelamite	18.20 C	1,200 lbs.	218.40
21"x 16"Gelamite	18.496 C	8,975 lbs.	1,660.02
15 "x 24" Gelamite	18.95 C	1,116 1bs.	211.49
35 1x 10"Gelamite	18.95 C	4,800 lbs.	909.60
E.P. 152	18.047 C	147,070 lbs.	26,542.76
E.P. 158	18.75 C	240,150 lbs.	45,028.19
E.P. 194	12.273 C	13,750 lbs.	1,687.65
Ammonium Nitrate	9.50 C	71,250 lbs.	6,768.75
Tritex	10.694 C	9,225 lbs.	986.53
Regular Primacord	32.50 M	32,123 ft.	1,043.99
Plastic Primacord	36.66 M	39,400 ft.	1,442.30
Connecting Wire	1.15 lb.	192 lbs.	220.80
XC-45 Boosters	50.00 C	650	325.00
17 M.S. Connectors	45.50 C	250	113.75
#6 Plastic Caps	15.00 C	100	15.00
Dry Fuse	5.697 C	1,427	8.13
Electric Caps	21.74 C	113	\$ 24.57
		a she had a start of the	\$ 87,216.88
Cost Per Ton of Mater	rial Broken		\$.118
	-9-		

HUMBOLDT MINE ANNUAL REPORT YEAR 1956 2

6. OPEN PIT: (Con't.)

c. Open Pit Mining:

Summary of Powder Used:

San Sand	Primary Blasting - Was	gon Drill Holes		and the second second
	Type	Unit Cost	Amount Used	Total Cost
	7/8"x 8"Gelamite	23.16 C	50 lbs.	\$ 11.58
N. C. Starter	14"x 8"Gelamite	18.35 C	780 lbs.	143.13
	2"x 8"Gelamite	18.20 C	275 lbs.	50.05
	21"x 8"Gelamite	18.20 C	250 lbs.	45.50
	27"x 16"Gelamite	18.455 C	4.600 lbs.	848.93
	2 ¹ / ₂ x 16"Gelamite	18.20 C	1.725 lbs.	313.95
	1불"x 24" Gelamite	18.95 C	3.619 lbs.	685.80
	Regular Primacord	32.00 M	26.465 ft.	846.88
	Dry Fuse	.625 C	24 ft.	.15
	#6 EB Cap	1.73 C	3	.05
State Barris	Electric Caps	21.95 C	13	2.85
1830 A.S.X				\$ 2,948.87
Sec. in	Secondary Blasting			
 A set that is the set of the se	7/8"x 8 Gelamite	19.90 C	10 lbs.	\$ 1.99
And and a second se	11"x 8"Gelamite	18.537 C	1,800 lbs.	333.67
Visit 1272 St	2"x 8"Gelamite	18.20 C	45 lbs.	8.19
1. Burger and rates	2 ¹ x 16"Gelamite	18.98 C	50 lbs.	9.49
	2 ¹ / ₂ "x 16"Gelamite	18.20 C	40 lbs.	7.28
	1를"x 24 Gelamite	18.523 C	65 lbs.	12.04
	Regular Primacord	30.728 M	2,675 ft.	82.20
	Friction Tape	.52 lb.	1/2 lb.	.26
1996年4月1日日間の	17 M.S. Connectors	45.50 C	100	45.50
	Cap Crimpers	2.15 Ea.	6	12.90
1. 1. 1. 1. 1. 1. 1. 1. 1. 1. 1. 1. 1. 1	Trucking, Servicing			6.58
				\$ 520.10

Summary of Footages Drilled:

Month	Jet Piercing	Wagon Drilling
Annil	2.064	769
May	3,701	5,302
June	2,721	4,912
July	a far a far an	
August	1,714	2,141
September	2,460	3,183
October	3,739	3,824
November	1,647	1,372
December	2,424	2,817
Total	20,470	24,320

HUMBOLDT	MINE
ANNUAL R	EPORT
YEAR .	1956

6. OPEN PIT: (Con't.)

c. Open Pit Mining:

Summary of Jet and 42" Diameter Drill Holes Blasted

	1	A Martin Star		Davidson	Care	Tona
_	No. of	Avg.	Average	Powder	Gros	Baal
Date	Holes	Deptn	Spacing	Factors	<u> </u>	ROCK
4-16	19	29.9	19 x 20	1.69	15,150	2,200
4-24	35	25.2	18 x 18	1.78	18,000	5,000
5-9	59	31.0	20 x 20	1.98	52,870	17,380
5-23	38	43.6	26 x 26	1.78	96,580	1994 - D. 19
6-15	27	43.8	26 x 26	1.75	53,590	16,080
8-20	42	42.2	17 x 25	1.36	47,410	18,210
8-23	6	34.2	20 x 16	1.03	5,800	
8-31	80	19.7	13 x 13	1.21	6,000	17,470
9-12	29	42.9	25 x 20	1.54	30,530	23,750
9-25	31	44.0	25 x 20	1.73	52,860	7,770
10-9	33	44.0	24 x 20	1.97	32,300	32,300
10-25	40	44.3	24 x 18	1.30	39,920	20,750
10-31	20	43.9	24 x 18	1.40	31,710	6,580
11-6	10	33.5	21 x 15	1.18	2,720	6,690
11-6**	43	25.0	9 x 9	1.00	8,000	8,000
12-7	55	28.1	18 x 17	1.36	36,625	4,125
12-21	31	39.5	18 x 18	1.02	35,320	5 - 2 - 2 - 2 - 2 - 2 - 2 - 2 - 2 - 2 -
NELCHARD					FLF DOF	104 205

Total 55

565,385 186,305

28

555 Jet Holes ** 43 $4\frac{1}{2}$ " Diameter Holes Drilled by $5\frac{1}{2}$ " Gardner-Denver 143 Wagon Drill

* Tons of Material Broken per Pound of Powder Used.

Average stemming - 12 feet. All drilling four feet below bench grade.

-11-

HUMBOLI	OT MINE
ANNUAL	REPORT
YEAR	1956

- 6. OPEN PIT: (Con't.)
 - c. Open Pit Mining:

Summary of Jet Piercer Operating Costs

Month	Fuel Oxygen <u>Oil</u>	Reamer Shells	Kelley	Burner Assembly	Piping	Misc. Open Supply	Misc. r. Maint. Supply	Royalty	Maint. Labor	Oper. Labor	Reamer	Total
March April	4.94 4199.01 533.00	25.02		163.07	.65 22.03	51.60 566.89	406.05	936.50	785.71	1501.92	280,13	1273.97
May June	10792.25 1699.14 9192.90 1782.66	272.17 123.86	107.90 471.90	150.00 166.35	104.40 375.18	427.90 765.00	744.53 549.52	2072.59 1604.69	372.17 627.51	2820.36 1930.48	402.20 457.28	19965.61 18047.33
August	4686.76 1206.27	253.36	219.65	200.62	220.09	833.83	327.59	1070.62	365.24	1636.34	265.34	11285.71
October November	10742.09 2111.16 3899.99 890.30	59.60	237.29 353.55	104.72	80.60	352.36	515.47	2412.07	378.08 714.06	3251.33	444.50	20609.68
December_	6804.57 1326.46	217.54	199.66	8.34	5.71	537.70	557.66	1032.28	1107.66	2025.52	248.20	14071.30
Total	57204.69 11222.92	1657.35	1747.00	949.36	1319.17	4403.01	4592.13	11987.96	4984.48	17464.38	2839.52	120371.97

Total Loaded Footage Jet Drilled Holes Total Drilled Footage Jet Drilled Holes Cost Per Foot of Loaded Footage Cost Per Foot of Drilled Footage Total Tons Blasted Ore and Waste Total Cost Per Ton Blasted - Ore and Waste

19,487' 20,470' 6.177 5.880 735,690 tons .164

-12-

HUMBOLI	OT MINE
ANNUAL	REPORT
YEAR	1956

7. PLANT:

a. General:

The crude ore utilized during the year was higher in grade than that processed in 1955. The net feed rate for the year was 111 L.T.P.H. The feed rate was limited somewhat throughout the year because of the extreme hardness of the crude ore.

The various rates were:

2 11	Crusher Feed	L.T.P.H.	Concentrator	Feed	L.T.P.H.	Concentrate	L.T.P.H.
2-1-1-	Gross Time	Net Time	Gross Time		Net Time	Gross Time	Net Time
1956	237.81	306.64	106.02		111.13	42.34	44.38
1955	221.90	300.53	86.50		98.60	31.59	36.00

The recoveries from the crude were:

	Crude		Co	oncentrate
1.20	Tonnage	% Fe	% Wt. Recovery*	% Fe Unit Recovery*
1956	518,044	29.70	37.42	77.94
1955	587,511	27.63	33.90	75.64

* Based on dry tons of feed and concentrate.

The operating time for the crushing plant and mill were:

	Crushing Plant	Mill
1956	77.55	95.40
1955	74.85	87.73

The outstanding mill delays were caused by mechanical failures in the pit and crushing plant which caused "out of feed" delays in the mill. These delays accounted for 58% of the total mill delay time. Operational type delays were responsible for 28% and mill mechanical failures accounted for the remaining 14% of the total delay time.

-13-
HUMBOLL	DT MINE
ANNUAL	REPORT
YEAR	1956

7. PLANT: (Con¹t.)

b. Monthly Production and Grade:

Month	Tonnage	% Fe	% P	% Si02	% S	% H20
April	9,749	61.57	.066	10.17	.010	7.30
May	28,063	61.57	.088	9.35	.014	7.05
June	34,156	61.85	.098	8.90	.040	7.22
August	19,227	61.55	.094	9.75	.013	6.39
September	30,795	62.18	.078	9.51	.011	6.03
October	29,636	62.24	.073	9.30	.012	6.07
November	29.478	61.81	.094	9.47	.023	6.46
December	24,000	61.71	.065	10.45	.008	6.95
Year	206,901	61.80	.084	9.50	.018	6.64

c. <u>Metallurgical Balance:</u>

Product	% Wt.	% Wt. Crude	<u>% Fe</u>	% Si02	Unit Recovery
Concentrate Flotation Tailing	39.19 <u>60.81</u>	37.42 58.06	61.86* <u>9.41</u>	9.48*	77.94 18.40
Flotation Feed	100.00	95.48	29.97	E. Carton	96.34
Secondary Cyclone Overflow	Real Contraction	4.52	24.08		3.66
Calculated Head		100.00	29.70		100.00

* Based on weighted averages carried at Humboldt Mill.

and the second	Iron	Phos.	<u>Sil.</u>	Sul.	Moist.
Concentrate Assay (Shipping Dep't.) (Dry) (Natural)	61.80 57.70	.084	9.50 8.87	.018	6.64
Guarantee for 1956 - (Dry)	61.80	.090	9.50	.010	7.50

-14-

HUMBOLI	DT MINE
ANNUAL	REPORT
YEAR	1956

7. PLANT: (Contt.)

d. Hourly Operating Rates:

Feed to Primary Crusher	Tons	of Operation	of Operation	(Gross)	(Net)
1956 Season	519,904	2186.25	1695.50	237.81	306.64
1955 Season	591,179	2652.50	1958.50	222.88	301.85
Ore in Process	1,728				
Ore for Roads	1,312	Sarah and Sarah	And And And And And	and the second	
Fine Ore Bin to Concentr	ator			Nga Zi	
1956 Season	518,044	4886.33	4661.51	106.02	111.13
1955 Season	587,511	6791.83	5958.42	86.50	98.60
Concentrates	a starter				11
1956 Season	206,901	4886.33	4661.51	42.34	44.38
1955 Season	214,532	6791.83	5858.42	31.59	36.00
Operating Time			THE SAL WE ST		19 5 1

Operating Time

1956	Season	95.40%
1955	Season	87.73%

Monthly Hourly Operating Rates:

1) <u>Long</u>	Tons Per Gross Hour	Feed to <u>Primary Crusher</u>	Fine Ore Bin to Concentrator	Concentrates
April		213.87	92.87	32.07
May	Cart Contract of Contract of Contract	264.93	113.12	40.49
June	Carly Construction of the Second	268.67	114.63	50.19
Augus	t	231.02	108.12	40.06
Septe	ember	301.34	110.92	46.38
Octob	er	260.88	99.60	41.11
Noven	iber	189.10	104.13	43.35
Decen	iber	188.90	98.28	36.14
	Year	237.81	106.02	42.34
2) Long	Tons Per Net Hour:			
April		288.21	97.58	33.70
May		306.84	116.08	41.55
June		326.48	116.31	50.93
Augus	t	305.11	114.44	42.40
Septe	ember	365.49	111.52	46.63
Octob	ber	340.07	106.95	44.15
Novem	iber	260.05	108.53	45.18
Decen	iber	267.35	110.90	40.79
	Year	306.64	111.13	44.38

HUMBOLL	DT MINE
ANNUAL	REPORT
YEAR	1956

7. PLANT (Contt.)

f.	<u>Pit - Crusher Time Distribution</u> :	Hours	Percent Delays	Percent Total Yearly Delay	Percent of Total Yearly Working Time
1)	Pit				
1	Pit Delay	167.75	100.00	34.18	7.67
2)	Crushers			34.18	7.67
in a start	A Company of the second state of the second state		and the second	S. S. S. S. S. Par	and the start
Pri	mary Chute Plugged	79.00	24.46	16.10	3.62
Chu	inks in Primary Crusher	53.00	16.41	10.80	2.42
Ter	tiary Crusher Repair	37.50	11.61	7.64	1.72
Pri	mary Crusher Jammed	26.50	8.20	5.40	1.21
Ros	s Feeder Repair	19.50	6.04	3.97	.89
#1	Buffer Belt	18.50	5.73	3.77	.85
#4	Conveyor	10.00	3.10	2.04	.46
#3	Conveyor	9.00	2.79	1.83	.41
Fee	d Box Plugged - Secondary Crusher	8.50	2.63	1.73	•39
Ins	tall Air Jets - Primary Chute	7.00	2.17	1.43	.32
#1	Screen	5.50	1.70	1.12	.25
Rep	air Secondary Crusher	5.50	1.70	1.12	.25
Tra	mp Steel in Secondary Crusher	5.00	1.55	1.02	.23
Ros	s Feeder Chute	5.00	1.55	1.02	.23
#2	Buffer Belt	5.00	1.55	1.02	.23
#1	Fines Chute	5.00	1.55	1.02	.23
Fir	e in Grid. Controller - Sec. Crusher	• 4.50	1.39	.92	.21
Pri	mary Crusher Chute Repair	4.50	1.39	.92	.21
#1	Conveyor	2.50	.77	.51	.11
Met	al Detector	2.50	.77	.51	.11
Sur	ge Bin Belt	2.50	.77	.51	.11
Sur	ge Bin Full	2.50	.77	.51	.11
Pla	nt Down-Approaching thunder storms	2.00	.62	.41	.09
#2	Conveyor	1.00	.31	.20	.05
Pow	er Failure	1.00	.31	.20	.05
Adj	ust Secondary Crusher	.50			02
	Total	323.00	100.00	65.82	14.78
	Grand Total	490.75		100.00	22.45



3:

HUMBOLI	TC	MINE
ANNUAL	RI	EPORT
YEAR		1956

7. PLANT: (Con't.)

g. Pit - Crusher Monthly Time Distribution:

Month	Hours Delay Time	Monthly Percent Operating Time	Monthly Percent Delay Time	Total Yearly Delays	Total Yearly Working Time
April	32.50	74.21	25.79	6.62	1.49
May	40.50	86.34	13.64	8.25	1.85
June	52.50	82.29	17.71	10.70	2.40
August	55.00	75.72	24.28	11.21	2.52
September	43.75	82.45	17.55	8.91	2.00
October	64.50	76.71	23.29	13.14	2.95
November	101.50	72.72	27.28	20.69	4.64
December	100.50	70.66	29.34	20.48	4.60
1.1.7. 9.00	490.75	77.55	22.45	100.00	22.45

h. Concentrator Delay Time:

1) Operational:	Hours Delay	Percent Delay	Percent Total Yearly Delay	Percent of Total Yearly Working Time
No feed - Tertiary Crusher Repairs	82.50	42.85	36.70	1.69
Startup and Shutdown - Repair Shift	26.20	13.61	11.65	.53
No feed - Ross Feeder Repairs	16.00	8.31	7.12	.33
No feed - Primary Chute Plugged	14.25	7.40	6.34	.29
Main Mill Breaker Overload	8.88	4.61	3.95	.18
Power Failure	8.16	4.24	3.63	.17
No feed - Pit Delay	7.92	4.11	3.52	.16
Hydroscillator Rakes Overloaded	5.92	3.07	2.63	.12
#3 Conveyor Motor Burned Out - No feed	5.67	2.94	2.52	.12
Thickener Overloaded	5.17	2.68	2.30	.11
Mill Pumps Sanded	4.50	2.34	2.00	.09
Flotation Machines Sanded	3.25	1.69	1.45	.07
No feed - Primary Crusher Jammed	3.00	1.56	1.33	.06
No feed - Fine Ore Bin Empty	.50	•26	•22	.01
Grinding Chutes Plugged	•47	.24	.21	-01
Feed Checks	.17	.09	.08	-
	192.56	100.00	85.65	3.94
2) Equipment:		Sec. Sec.		
Mill Pumps	16.19	50.19	7.20	.33
Thickener Drive Thrust Bearing Failure	7.91	24.52	3.52	.16
Hydroscillator Rake Repair	4.41	13.67	1.96	.09
Completion of Ball Mill Relining Job	2.50	7.75	1.11	.05
Hydroscillator Bowl Repair	1.25	3.87	.56	.03
	32.26	100.00	14.35	.66

Total

-17-

224.82

100.00

4.60

HUMBOLI	DT MINE
ANNUAL	REPORT
YEAR	1956

和行

CINE

7. PLANT: (Cont.)

C COLLOM ERE

IENL BOMD

。同的些国际的时候

i. Concentrator Monthly Delay Time:

Month	Hours Delay 	Monthly Percent Operating Time	Monthly Percent Delay Time	Percent of Total Yearly Delays	Percent of Total Yearly Working Time
April	14.67	95.17	4.83	6.53	.30
May	17.63	97.46	2.54	7.84	.36
June	9.83	98.56	1.44	4.37	.20
August	26.50	94.48	5.52	11.79	.54
September	3.53	99.47	.53	1.57	.07
October	49.50	93.13	6.87	22.02	1.01
November	27.58	95.94	4.06	12.27	.57
December	75.58	87.73	11.38	33.61	1.55
Year	224.82	95.40	4.60	100.00	4.60

j. Monthly Rod, Ball and Reagent Consumption:

Month	Rod	s	Ball	s	Reage	nts
	#	#/ton keed	#	#/ton	#	#/ton
April	74,600	2.642	28,300	1.002	26,884	.952
May	116,809	1.490	60,600	.773	98,622	1.258
June	98,451	1.262	57,560	.725	144,415	1.851
August	73,150	1.410	47,850	.922	72,255	1.391
September	129,560	1.759	62.045	.842	112,846	1.532
October	128.835	1.794	71.660	.998	131,044	1.825
November	80,990	1.144	81.055	1.145	120,394	1.700
December	80,035	1.226	52,000	.797	70,971	1.088
	782,430	1.510	461,070	.890	777,431	1.501

-18-

HUMBOLDT MINE ANNUAL REPORT YEAR

7. PLANT: (Con't.)

k. Plant Testing:

Plant testing was continued on a limited basis during the year. Changes in equipment, reagents and grinding media were made to effect immediate improvements as well as to continue a study of certain problems on a long-range basis.

The new Denver flotation machines, which were installed during the idle period in 1955, were placed in operation in 1956. As originally furnished, the rougher cells were continually overloaded and produced a tailing which was relatively high in iron content. After several months of operation and numerous minor adjustments, the Denver Equipment Company furnished Type M mechanisms to replace the Sub-A mechanisms that were originally installed in the rougher cells. After the installation of the Type M mechanisms in the roughing section, this machine produced a much better tailing. The Type M mechanism also drew much less power and, therefore, it becase possible to increase the feed rate to this machine without overloading the motors.

The high consumption of Acintol #2 was under consideration throughout the year. This abnormal consumption was apparent only when treating knob one ore. When treating knob three ore, the reagent consumption was normal as compared to 1955 when a Neofat 139-Red Oil mixture was used as the primary collector. In December, an Acintol #2-Neofat 139 mixture was used for a limited time on knob one ore. This mixture reduced the actual consumption by one-third and slightly better metallurgical results were obtained. On knob three ore, which is leaner and often argillaceous, little difference in results or consumption was noted. As far as costs were concerned, the reduction in reagent quantities was offset by the higher cost of Neofat 139. Plant testing on knob one ore with the Neofat 139-Tall Oil mixture will be continued in 1957 to establish more definite information.

In December, a car load of Pamak 1, produced by the Hercules Powder Company, was tested. The Pamak 1 was similar in composition to Acintol #2 and competitively priced. Metallurgically speaking, there was little or no difference found in the two tall oil products. In the future, Pamak 1 will be used at Humboldt instead of Acintol #2.

In the grinding circuit, changes were made in the size of the grinding rods. Grinding rods measuring 3", $3\frac{1}{2}$ " and 4" were charged into the rod mills at various times during the year. Indications are that a higher circulating load in the grinding circuit is obtained with the larger rods. Accordingly, it is now planned to continue the use of the larger rods along with the addition of some $1\frac{1}{2}$ " balls to the normal ball charge of $1\frac{1}{4}$ " balls. A charge of 15% of $1\frac{1}{2}$ " balls and 85% of $1\frac{1}{4}$ " balls will be used along with the larger rods. The advantages gained by using the larger rods are: 1) a better wear pattern on the rods and subsequent lower rod consumption and 2) a slight gain in grinding capacity.

-19-



8. MAINTENANCE, REPAIRS AND CHANGES:

a. Crushing Plant:

In addition to routine repairs in the crushing plant, the following major repairs and plant changes were completed in 1956:

- April A new hydraulic reset system was installed on the Hydrocone crusher.
- 2) August The eccentric sleeve and bottom shell bushing was replaced on the primary crusher.
- 3) October The eccentric sleeve was replaced on the Hydrocone crusher.
- 4) November The splitter beam was removed and the dead bed slide dumping arrangement, into the primary crusher was completed.
- November The Roto-Clone dust collector unit in the secondary building was put into operation. The entire dust collection system is not complete.
- 6) December New drive gears on the Ross Feeder were installed to replace broken gears.
- 7) December All sixteen of the top shell studs in the Hydrocone crusher were replaced. Thirteen of the sixteen studs broke on December 4th.

b. Concentrator:

Major repairs in 1956 were:

- 1) The heads and rotors in both of the vacuum pumps were replaced.
- 2) A new oscillator was installed on the hydroscillator.
- 3) Larger feed and equalizer lines were installed on the filtrate pumps.4) A new pinion gear was installed on both rod mills.
- 5) The thickener drive thrust bearing was replaced four times during the
- year because of the inadequate design of the thrust bearing assembly.6) The ball mill was completely relined. This was the first time the cylinder liners were replaced since the beginning of operations at
- Humboldt Mine.
 7) The eight Sub-A mechanisms in the rougher section of the new Denver flotation machines were replaced with Type M mechanisms.

AN FUNCT

-20-

HUMBOLDT MINE ANNUAL REPORT YEAR 1956

9. E & A's:

Construction and Stripping E & A's:

	and the second	1956 Expenditures	Total Expenditures
	HM-6 Drop Ball	\$ 64.832.21	\$ 64.832.21
	HM-9 Plant Changes	3.646.22	40.321.66
	HM-11 Dust Collection	21.328.51	21.328.51
	HM-12 Stripping	16,982,16	95.400.00
	HM-13 Tractor	22.351.00	22.351.00
	HM-16 Air Trac Drill	1.757.88	4.757.88
	HM-17 Pit Pump	962.59	962.59
	HM-18 Power Lines	6.313.11	6.313.11
	HM-20 Control Center	3,1,97,81	3,197,81
	HM_21 Plant Improvement	1.821.22	1.821.22
	HM_22 Strinning	11.7.71.2.51	1),7,7),2,5)
	HM_23 Dick-up Truck	1.878.64	1.878.64
	HM-25 Book Box	2,165,61	2,165,61
	UM 26 21 Vd Bucket	1. 336 1.1.	1,336.1.1
	IM 27 Durne	3 178 80	3, 178, 89
	IN 20 Primary Cruching Plant Improvements	557 38	557 38
	IN 20 MD 18 Proster	2 000 00	2 000 00
	HM-50 ID-10 Iractor	8 1.10 51	8 1.70 El.
2	HM-35 Share of Costs - Mot Filot Flant	206 12	206 12
1	Hu-jo Freiminary Design and Fiant Expansi	200.12	EL2E 071 1E
	TOTAL	\$313,300.01	\$455,014.15
	A REAL PROPERTY AND A REAL		

10. COST OF PRODUCTION:

Comparison: a. 1956 1955 \$1.915 .676 2.482 \$1.967 Pit Expense .633 Crushing and Screening Milling Expense Tailings Disposal Stocking Expense General Expense 2.230 .027 .070 .714 .039 .029 .050 .608 .027 Crude Charged to Operating .068 .094 Miscellaneous .119 .097 Taxes .160 Other Expenses and Income 1.031 .968 Depletion and Depreciation Shipping Expense .085 .086 \$7.011 \$7.346* Cost at Mine

* Includes W&I Costs

HUMBOLDT MINE
ANNUAL REPORT
YEAR 1956
CONTRACTOR AND A CONTRACTOR OF A DECK

10. COST OF PRODUCTION:

39

b. <u>Detail of 1956</u>:

	April	May	June	July	Aug.	Sept.	Oct.	Nov.	Dec.	Total
Pit Expense	2.669	1.997	1.672		1.885	1.960	1.935	1.681	2.607	1.967
Crushing & Screening	.700	.619	.425	-	.631	•567	.624	.631	.975	•633
Milling Expense	3.276	2.329	1.805	-	2.219	2.069	2.387	2.114	2.208	2.230
Tailings Disposal	.052	.036	.019		.022	.052	.023	.019	.020	.029
Stocking Expense	.090	.077	.030		.044	.024	.053	.022	.101	.050
General Expense	1.062	.502	•450	-	.862	•543	.462	.307	1.245	.608
Crude Chg ¹ d. to Oper.					.291	- 3	-		-	.027
Miscellaneous	.010	.103	.040	1997 - 199	.106	.113	.056	.122	.173	.094
Taxes	.297	.103	.085		.121	.094	.065	.065	.091	.097
Other Exp. & Income	.309	.158	.155	19 - A	.190	.164	.152	.100	.178	.160
Depl. and Deprn.	1.158	.938	.899	-	•975	.901	1.354	1.199	.961	1.031
Shipping Expense	.016	.031	.050	1994	.160	.202	.084	.1.04	<u>-</u>	.085
TOTAL COST	9.639	6.893	5.630		7.496	6.689	7.195	6.364	8.559	7.011
Production *Includes 1.797 tons	9,749	28,063	34,156		19,227	30,795	29,636	29,478	24,000	206,901*

January, February and March - Winter & Idle Expense July - Idle Expense

-22-

HUMBOLDT MINE ANNUAL REPORT 1956 YEAR

10. COST OF PRODUCTION:

c. Analysis of Costs:

The total cost of producing concentrate at the Humboldt Mine during 1956 again was lower than the preceding year. The yearly decline since the opening of the mine is shown in the following table:

Year	Cost of Production	Total Cost
1954	\$ 7.383	\$ 8.754
1955	5.869	7.123
1956	5.638	7.011

To permit an accurate comparison, the 1955 figures in the above table are adjusted from these shown on cost sheets because of idle charges.

Although the life of the Humboldt operation has extended over three years, the plant, at the end of 1956, had actually run only 27¹/₄ months. This comparatively limited operating time, accompanied by idle-periods and an almost continuous program of changes, has greatly affected overall costs in each of the years. Of even greater consequence, has been the cost of mining because of the complex geology and rugged terrain. Mining costs will undoubtedly show the greatest improvement in the future when the irregular knobs are removed and stripping has progressed to the extent that more elaborate mining plans can be used to off-set the geological problems.

During 1956, two general factors affected operating expenditures and prevented a larger decrease in the cost of production. A large portion of the pit and mill equipment had reached a point where replacement parts were required which increased maintenance costs over 1955 by at least \$11,000 per month. It is expected that this cost will level off and not increase over the 1956 total. Secondly, mining was conducted in the first knob area which, because of joints, fractures and dikes, tended to fragment in large chunks when blasted. Although experimentation with hole spacings and powder charges improved the situation slightly, mining and crushing costs were, in general, affected by the need for utilizing this ore during most of the operating months in 1956. Of a more detailed nature, but also adding to the 1956 costs, were high rental charges, as five new trucks, one shovel and one tractor were operated on this basis. Referring again to the geological problems of the pit, waste removal charged to the ore operation, involving dikes, lean and unoxidized iron formation within the ore body also added to mine expenditures during 1956.

Since most of the above items that have been affecting costs to date at the Humboldt operation should be largely eliminated in the future, it is reasonable to estimate that a steady improvement to the Humboldt profit margin will be realized over the coming years.

-23-

HUMBOLI	DT MINE
ANNUAL	REPORT
YEAR	1956
1 100 100 100 100 100 100 100 100 100 1	

11. TAXES:

Description	<u>_1956</u> Valuation	Taxes	_ <u>195</u> Valuation	5 Taxes
Humboldt Mine, including stockpile, supplies and equipment as placed by				
State Mine Appraiser:	* *** 000 4	17 210 00	¢ 050 000	\$17 000 00
Real Astate	\$ 002,000 \$	71 600 00	¢ 050,000	\$ 200.00
Personal Property	730,000	14,000.00	410,000	0,200.00
Dr. Burke Camp	400	8.00	_ 400	8.00
	The second	31,848.00	and share the	25,208.00
Collection Fee	Low Manager	318.48	a la la la	252.08
TOTAL HUMBOLDT MINING CO.	\$1,592,400	32,166.48	1,260,400	\$ 25,460.08
Tax Rate	\$20.	.00	\$20	.00

41

Nature Days Compensation Paid Lost Name Date of Injury of Injury 40 336.00 Broken bone George Auge June 18 Right hand Laceration - 3rd finger 25 259.45 August 21 George Karvela Right hand 390.00 42 Strained back Norman Bradbury August 31

13. PROPOSED NEW CONSTRUCTION:

The principal construction change definitely scheduled for 1957 is a remodeling of the primary crushing building to facilitate the installation of a pan feeder under the crusher. This work involves mainly concrete and steel changes in the area between the crusher discharge and the tail end of conveyor #1.

Also planned for 1957 is the extending of the concrete floor in the mill. This project will help solve the clean-up problem associated with ball mill #2 as the floor in this section of the mill has never been finished.

HUMBOLDT MINE ANNUAL REPORT YEAR 1956. SS: @ MA I ADIT

14. EXPLORATION:

An exploration diamond drilling program was initiated during the latter part of December. Crews moved into the area and set up on DDH #8 (Sec. 2), located at co-ordinates 4809.09 S. and 8394.90 W. The hole is being drilled on a course of S. 38° 46' E. at an inclination of -45° from an elevation of 1541.83'. At the close of the year, 68' of clear, sandy overburden plus 8' of broken quartzite ledge and boulders had been penetrated. It is hoped that this hole will provide pertinent information relevent to (1) the location of the hanging wall contact, (2) the magnitude of the extension of iron form-ation as indicated by DDH #7 (Sec. 2), (3) the nature of the iron formation present including some indication as to the degree of contamination by dikes and lean zones, (4) the interpreted fault zone possibly relating its general trend and (5) the mining footwall contact.

In conjunction with diamond drilling, a program of detailed geophysical exploration has been set up for the coming year. The survey will initiate with north-south lines at the southwest portion of the Weber Lease and continue eastward until anomilies indicate it no longer possible.

Cursory field examination indicated the possibility of finger-like protuberances of oxidized iron formation beyond the present mining footwall limits. A thorough plane table mapping project has been proposed for the coming field season. Although outcrops are scarce, certain areas may appear fruitful meriting a few short drill holes.

15. EQUIPMENT RECEIVED AND PROPOSED NEW EQUIPMENT:

- a. Equipment Received:

 - $1 2\frac{1}{2}$ Yard Dipper 1 2000 GPM Pit Pump
 - 2 200 GPM Pit Pumps
 - 1 F-250 Pick-up Truck
 - 1 MG Set for Grind Section
 - 2 400 Amp. Welders
 - 1 Air Trac Wagon Drill
 - 1 D-7 Tractor
 - 1 1200 GPM Pit Pump
 - 1 Roto Clone Dust Collector
 - 1 Multi Clone Dust Collector
 - 1 5" Hazelton Pump
 - 1 Control Center for Flot Section
 - 1 Hydrocone Reset System
 - 1 Oscillator Bowl

- b. Proposed New Equipment:
 - Pioneer Pan Feeder
 - 13 Flotation Tanks
 - 1 Merrick Weightometer
 - 1 Hough Payloader
 - 1 #143 52" Air Trac Drill 1 TD-18 Tractor

 - 1 Set of Shells for Tertiary Crusher

42

- 1 Pick-up Truck

-25-

1. GENERAL:

Production for the Mather Mine, "A" Shaft for the year was 1,252,192 tons. This is the highest annual tonnage produced to date at the "A" Shaft. This production was achieved in spite of a five week shut-down due to the steel strike. Operations were on a three shift five and one-third day per week schedule until November 11th when a five day per week schedule went into effect. Shipments from the pocket commenced on April 6th and continued through November 26th and shipments from the stockpile commenced on April 9th and continued through December 3rd. A total of 1,118,694 tons were loaded out during that period.

The analysis of the Mather Mine, "A" Shaft product for the year was 58.33 Fe., 9.02 Si., and .017 Sul. The analysis of the shipments for the year was 52.543 Fe. Natural. This analysis compares most favorably with the analysis in 1955 when the shipments were 51.45 Fe. Natural.

Notwithstanding the five week production loss because of the steel strike and the increase in the cost of labor and supplies, the cost of production decreased \$.037 per ton as compared to 1955. The total cost at the mine including depreciation, taxes, and loading and shipping costs showed a decrease of \$.160 per ton. The tons per man per day increased from 9.17 in 1955 to 10.04 in 1956, another new high for the Mather Mine, "A" Shaft.

The steel shortage continued during the year and necessitated substitutions in structural steel fabrication. The Bethlehem yielding steel arches were introduced as supports during the year and were found to be very satisfactory.

The underground drilling program for outlining new ore reserves continued throughout 1956. Diamond drills operated from the 7th, 8th and 9th Levels with the emphasis of the drilling being placed on outlining the reserves between the 8th and 9th Levels and exploring below the 9th Level. New ore reserves disclosed were not sufficient to offset the production during the year, with the result being a net loss in reserves of 587,641 tons.

Labor relations for the year were satisfactory. The steel strike went off quietly and maintenance men needed for service work were allowed to enter the property. There was one formal grievance presented with respect to the filling of a job vacancy. The union dropped this grievance in the 4th step.

Mining operations were conducted principally on the 7th and 8th Levels with the 5th and 9th Levels contributing a minor percentage of the production. By levels, production was as follows: 5th Level - 26,490 tons or 2.13% of the total product; 7th Level - 598,787 tons or 48.15% of the total product; 8th Level - 611,792 tons or 49.20% of the total product; and 9th Level - 6,486 tons or .52% of the total product.

Main level drifting was completed on the 9th Level and cross-cut development will continue into 1957. All cross-cut development was completed on the 8th Level during the year. The main conveyor belt project on the 9th Level was about 95% complete at the end of the year and this unit will go into operation early in 1957.

One sub-level conveyor went into operation on the 8th Level during the year. Excava-

-1-

Mather "A" 1956

1. GENERAL: (Continued)

tion for a second sub-level conveyor on 8th Level and one on the 9th Level was completed. The installation work on these additional units will carry over into 1957.

The development of ore reserves below the 9th Level at the Mather Mine, "A" Shaft was begun recently to make available the ore between the 12th and 9th Levels. Because of the length of time required to reach a depth three levels below the present lowest operating level, and then to bring the new levels into production, it was important that the development be started at the earliest possible date.

After careful consideration of all the factors involved, it was decided that a single inclined belt conveyor for handling ore and rock, together with the extension of the shaft (cage, ladder, and pipe compartments only) for handling men and supplies, would provide the best solution to the problem. At each level below the 9th, only the cage plat need be cut and no skip loading pockets at the shaft would be required.

Plans for the actual development of the inclined drift and the installation of the belt conveyor are quite unique. After driving the first 500 to 800 feet of the belt drift by the conventional slusher method, a portion of the belt conveyor will be installed. The tail pulley section will be an integral part of a scraper slide, which can be moved down the inclined drift as required. The scraper slide will protect the belt during the blasting operation and also provide a simple method for loading out the development rock. Additional belting will be spliced in each time the scraper slide is advanced.

Because of the extremely long inclined drift, it is planned to install a single drumshaft type hoist to handle men and supplies through this drift during the development period. The hoist will operate a car on a standard 30" gauge track. The track will be installed alongside of the belt conveyor as development work progresses.

After the conveyor drift is completed a crushing and feeding station similar to the arrangement on the 7th and 9th Levels will be installed.

During the mining operation, ore will be transferred by tram cars and sub-level conveyor belts to a main storage trench located adjacent to the crushing and feeding station. The ore will then be scraped from the trench over a scalper screen, with the oversize material discharging into a jaw crusher and the undersize to a pan feeder, which in turn will load the belt. The material will travel 3700 feet along the 36" conveyor belt to the head pulley end where it will be discharged into a raise leading to the $9\frac{1}{2}$ Level skip loading station.

By use of the inclined conveyor system the lower operating levels will come into production in a minimum length of time and the usual skip plats, pockets, and shaft skip compartments will be eliminated below the present shaft bottom.

Mather "A" 1956



2. PRODUCTION:

a. Production by Grade and Months:

		Stockpile		1955
Grade:	Product	Overrun	Total	Total
Mather	1,243,555	8,637	1,252,192	
Mather Special	-	-	-	2 015 100
Total	1,243,555	8,037.	1,252,192	1,045,409
Rock			57,860	67,705

Months:	Ore	Rock
January	100,857	5,533
February	107,957	5,335
March	109,430	7,678
April	112,699	8,118
May	126,503	4,400
June	112,719	2,376
July		
August	111,436	4.664
September	121,116	5.962
October	141.948	6.523
November	112,314	3.146
December	95,213	4.125
Total	1,252,192*	57,860

* Total includes 8,637 tons, current year stockpile overrun pro-rated monthly.

b. Shipments:

	Pocket	Stockpile	Total	1955 Total	Decrease
Mather	699,378	419,316	1,118,694	1,415,286	and the second
Mather Special	699,378	419,316	1,118,694	1,472,005	353,311

A balance of 230,325 tons of ore was left in stock as of the end of the year. An overrun of 8,637 tons was developed from the north and center stockpiles which were completely loaded out at the end of the shipping season.

2. PRODUCTION: (Continued)

c. Ore Statement:

		Mather		1955
	Mather	Special	Total	Total
On Hand January 1, 1956	96,827	And a second	96,827	523,423
Output for Year	1,243,555	10 17 17 10 19 19 19 19 19 19 19 19 19 19 19 19 19	1,243,555	1,037,680
Transfers				-
Overruns	8,637	States - Connega	8,637	7,729
Total	1,349,019	-	1,349,019	1,568,832
Shipments	1,118,694	-	1,118,694	1,472,005
Balance on Hand	230,325		230,325	96,827
Decrease in Output				
Increase in Output			206,783	130,004
Decrease in Ore on Hand				426,596
Increase in Ore on Hand			133,498	

Working Schedule:

- 1956 3-8 hr. shifts, 5-1/3 days per week, Jan. 1st to Nov. 11th. 3-8 hr. shifts, 5 days per week, Nov. 11th to Dec. 31st.
- 1955 2-8 hr. shifts, 4 days per week, Jan. 1st to April 18th. 2-8 hr. shifts, 5 days per week, April 18th to Aug. 1st. 3-8 hr. shifts, 5-1/3 days per week, Aug. 1st to Dec. 31st.
- 1954 3-8 hr. shifts, 5 days per week, Jan. 1st to March 1st. (Excluding a small production crew, Saturday, day shift.) 3-8 hr. shifts, 5 days per week, March 1st to April 5th. 3-8 hr. shifts, 4 days per week, April 5th to May 15th. 2-8 hr. shifts, 4 days per week, May 15th to Dec. 31st.
- 1953 3-8 hr. shifts, 5 days per week, Jan. 1st to Dec. 31st. (Excluding a small production crew, Saturday, day shift.)
- 1952 3-8 hr. shifts, 5-2/3 days per week, Jan. 1st to May 31st. 3-8 hr. shifts, 5-1/2 days per week, June 1st to Sept. 30th. 3-8 hr. shifts, 5-1/3* days per week, Oct. 1st to Dec. 31st. (*Excluding a small production crew, Saturday, afternoon shift.)

-4-

Mather "A" 1956

2. PRODUCTION: (Continued)

d. Division of Product by Levels and Months:

Months	Fifth (2050') Level	Seventh (2400') Level	Eighth (2590') Level	Ninth (2810') Level	Total	Rock
January	425	51,750	47,700	175	100,050	5,533
February	1,762	54,975	49,488	925	107,150	5,335
March	1,550	60,875	44,675	1,475	108,575	7,678
April	5.073	73.640	32,102	1,077	111,892	8,118
May	4,872	68,862	51,869	57	125,660	4,400
June	3.341	48,953	59,253	413	111,960	2,376
July				- A. A.		1001 - 100-173
August	1,907	44,888	63,845	85	110,725	4,664
September		47,732	72,530	95	120,357	5,962
October	100	57.980	83,101		141,081	6,523
November	935	46,569	62,654	1,397	111,555	3,146
December	<u>6,625</u> 26,490	42,563	44,575	787 6,486	<u>94,550</u> 1,243,555	4,125
		the state of the second st				

Current Year Stockpile Overrun

Carl La Part

8,637

e. Production Delays:

Delays due to mechanical or electrical difficulties during the year were few and minor, with no delays of more than a single shift duration.

During the month of June a loss of five operating shifts was caused by the Mather Mine, "B" Shaft underground fire.

The industry wide steel strike interrupted operations from July 2nd to August 6th. Work was resumed on August 7th.

-5-

3. ANALYSIS:

a. Average Mine Analysis on Output:

Grade	Iron	Phos.	Silica	Sulphur
Mather	58.33	1.11.20.20	9.02	.017

b. Average Analysis of Shipments:

Grade	Iron	Phos.	Silica	Mang.	Alum.	Sulphur	Lime	Mag.	Loss	Moist.
Mather	58.20	.094	9.27	•33	3.08	.015	•53	.99	1.85	9.72

c. Average Analysis of Ore in Stock: (Natural)

Grade	Tons	Iron	Phos.	Silica	Mang.	Alum.	Sulphur	Lime	Mag.	Loss	Moist.
Mather	230,325	52.615	.085	8.16	.30	2.78	.019	.48	.89	1.67	9.72

-6-

4. COST OF OPENING, EQUIPPING, DEVELOPING AND OPERATING:

Expenditures under E&A's amounted to \$871,264.87. The following table shows the main items of capital expenditures which are included in the total charges above.

MANY SAL

同时回题 上周之间

-7-

SHELLIOLLION % SK

	Percentage E&A Charges	of Total in 1956
Main Level Development	42.2	
Underground Conveyor Belt and Crusher	2). F	
Underground Exploration	5.7	
Underground Equipment	20.1	
Surface Projects	2.0	
Subsidence	5.5	

Mather "A" 1956

4. <u>COST OF OPENING, EQUIPPING</u>, <u>DEVELOPING AND OPERATING</u>: (Continued)

Comparative Mining Costs:

The Cost of Production of \$3.670 showed a decrease of \$.037 per ton as compared to 1955. The decrease in cost was achieved despite an increase in the cost of labor and supplies.

The results achieved in keeping the cost down were a reflection of the increase in unit production from 9.17 tons per man per day in 1955 to 10.04 in 1956.

The Total Cost at Mine was \$.160 less than 1955.

Both the Cost of Production and Total Cost at Mine were the lowest they have been since 1950.

		1956	1955
Product		1,252,192	1,045,409
Underground Ca Surface Costs General Mine I Cost of Prod	osts Expense duction	2.859 •359 <u>•452</u> 3.670	2.862 .347 <u>.498</u> 3.707
Depreciation:	Pre-Production Development Plant & Equipment Movable Equipment Development Miscellaneous Equipment Amort. of Defense Facilities Current Year's Development Adjustment Prior Year's Charges	.013 .186 .004 .103 .001 .043 .373	.013 .162 .007 .104 .001 .096 .494 .090
Taxes Administration Loading and Sh	n nipping	.245 .050 .054	.283 .050 .075
Total Cost a	at Mine	4.742	4.902
Budget - Cost Budget - Total	of Production L Cost at Mine	3.983 4.684	3.969 4.655
Number of Shit	fts and Hours	33 1-8 hr. 36 2-8 hr. 204 3-8 hr.	22 1-8 hr. 136 2-8 hr. 97 3-8 hr.
Total 8 Hour (Number of Open	Operating Shifts rating Days	717 239	585 240-2/3
Average Daily	Product	5,239	4,344

Mather "A" 1956

4. COST OF OPENING, EQUIPPING, <u>DEVELOPING AND OPERATING</u>: (Continued)

Proportion of	Labor and Su	oplies
\$3,181,749.53	2.541/ton	59%
2,235,652.13	1.785/ton	41%
\$5,417,401.66	*4.326/ton	100%
	Proportion of \$3,181,749.53 	Proportion of Labor and Sug \$3,181,749.53 2.541/ton 2,235,652.13 1.785/ton \$5,417,401.66 *4.326/ton

*Does not include Amortization of Defense Facilities and Allowance Under Section 616.

-9-

Mather "A" 1956

4. <u>COST OF OPENING, EQUIPPING</u>, <u>DEVELOPING AND OPERATING</u>: (Continued)

Detailed Cost Comparison:

8

	L,	150	19	755	
	Amount	Per Ton	Amount	Per Ton	
Underground Costs:					
Development	834.977.79	.667	635.272.64	.608	
Mining	1.384.713.19	1,106	1,142,013,01	1.094	
Tramming	586 317.37	1.68	504 120,48	.1.82	
Power Adjustment	4. 539.86	.003	Jud , 1	•***	
Ventilation	13 1.1.0 50	011	20 096 70	010	
Dumping	50 135 00	01.9	56 857 06	055	
Company and tim Times	56 102 10	.040	15 250 81	.013	
Compressors and AIF Lines	20,473.10	.045	47,270.04	.043	
crushing and Screening - UG	30,033.33	.024	23, 319.10	.022	
Maintenance:	194,365.75	.155	195,057.01	-186	
Pocket and Chutes	10,240.39	.008	6,994.78	.006	
Mining Equipment	128,754.20	.103	119,299.94	.115	
Levels and Cross-cuts	43,709.33	.035	42,687.95	.041	
Shaft	4,433.66	.003	8,990.44	.008	
Telephones and Safety Devices	48,740.04	.039	42,407.24	.040	
Vacation Pay	141,855.19	.113	101, 514.35	.097	
Holiday Allowance	46.476.51	.037	48,106.79	.046	
Fire Loss	365.16	.000			
Total Underground Cost	3,580,150.74	2.859	2,991,988.33	2.862	
Surface Costs:		Carl Harris			
Hoisting	167.029.41	.133	112,987.64	.110	
Crushing and Screening - Surface	14.335.89	.012	12,991.34	.012	
Stocking	65.227.81	.052	43.730.66	.042	
Timber Yard	61, 527,46	.049	47.948.17	.045	
Dry House	12 097.04	.033	42.665.52	.041	
Policing	19 755 51	.016	19,106,55	.018	
Conomal Sumface	17 070 70	014	21, 171 89	.023	
Maintenance:	1,010.17	.014	~+, 1/1.07	••••	
Headframe Bldg. and Equipment	12.057.94	.010	7,936.35	.007	
Other Mine Buildings	7.096.99	.005	17,504.08	.017	
Telephones and Safety Devices	1.389.61	.001	1.018.47	.001	
Vacation Pay	35.463.78	.028	25.378.59	.024	
Holiday Allowance	6,676.64	.006	7,303.07	.007	
Total Surface Cost	449,728.90	•359	362,742.33	.347	
General Mine Expenses:					
	0.0// 33	000	10 110 04	000	
Geological Department	9,900.11	.008	10,119.20	.009	
Mining Engineering Department	39,740.45	.032	37,424.47	.036	
Mech. Engineering Department	6,725.08	.005	8,354.45	.008	
Safety Department	10,541.34	.009	8,995.65	.009	

Mather "A" 1956

8 1. 52

-10-

COST OF OPENING, EQUIPPING, DEVELOPING AND OPERATING: (Continued)

4.

Detailed Cost Comparison: (Continued)

	1956		1955	
	Amount	Per Ton	Amount	Per Ton
General Mine Expenses: (Continued)				
Research Laboratory	5,704.54	.005	16,120.87	.015
Analysis and Grading - Laboratory	41,433.41	.033	40,498.50	.039
" " " - Shipping	7,363.00	.005	6,489.31	.006
Special Expense - Pensions	142.67	.000	146.42	.000
" " - Retirements	6,384.86	.005	7,169.75	.007
" " - Hygiene Clinic	9,119.20	.008	8,856.51	.008
" " - Employment Office	1,693.43	.001	1,602.60	.001
Ishpeming Office	122,642.99	.098	107,519.48	.103
Mine Office - Supt. and Clerks	71,136.08	.057	63,540.16	.061
Central Warehouse Overhead	28,884.75	.023	24,001.48	.023
Insurance - Property	4,195.85	.003	4,309.51	.004
" - Group, Health and Life	54,019.11	.044	27,148.83	.026
" - Group Annuity	15,610.60	.012	12,224.42	.012
" - Catastrophe	6,414.71	.005	4,749.00	.005
Personal Injury - Comp. & Doctors	34,624.30	.028	52,605.24	.050
" " - Comp. Department	6.91	.000	A LANGE MELETING	
Taxes - Unemployment Insurance	25,467.30	.020	18,078.75	.017
" - Old Age Benefit	50,492.09	.040	47,023.36	.045
Electrical Engineering Department	2,790.69	.002	2,948.84	.003
Employees Insurance & Compensation	11,372.56	.009	10,849.67	.011
Design Department	67.66	.000		
Total General Mine Expenses	566,539.69	•452	520,776.53	.498
COST OF PRODUCTION 4,	596,419.33	3.670	3,875,507.19	3.707

COST OF OPENING, EQUIPPING, DEVELOPING AND OPERATING: (Continued)

4.

Detailed Cost Comparison: (Idle Expense Due to Strike)

The industry wide steel strike interrupted operations from July 2nd to August 6th. Work was resumed on August 7th.

	Amoun
Inderground Costs:	
Development	1.220.2
Mining	2.379.7
Tramming	8,603,6
Ventilation	1,120,2
Dumping	3 601 1
Compressions & Aim Lines	2 003 0
Compressors & Air Lines	2,775.0
Underground Curperintendence	\$ 035 \$
Maintenence	0,055.0
Maintenance:	202.0
Pockets and chutes	650 0
Mining Equipment	6 000.9
Levels and Cross-cuts	0,839.3
Telephones and Safety Devices	2, 522.2
Holiday Allowance	
Total Underground Cost	38,470.8
ourface Costs:	
Hoisting	5,275.3
Timber Yard	96.3
Dry House	1,280.4
Policing	2,118.8
General Surface	537.9
Telephones and Safety Devices	294.4
Holiday Allowance	148.2
Total Surface Cost	9,751.6
eneral Mine Expenses:	
Geological Department	1,198.7
Mining Engineering Department	3,010.2
Mech. Engineering Department	516.5
Safety Department	1,008.0
Research Laboratory	184.1
Analysis and Grading - Laboratory	1.201.9
" " - Shipping	732.0
Special Expense - Pensions	17.0
I II _ Retirements	796.0
II II _ Hydiana Clinic	3/.8.7
" - nygtene utille	192 0
- Embroyment Olifice	172.0

Mather "A" 1956

COST OF OPENING, EQUIPPING, DEVELOPING AND OPERATING: (Continued) 4.

Detailed Cost Comparison: (Idle Expense Due to Strike) Continued

Amount

General Mine Expenses: (Continued)	
Ishpeming Office Mine Office - Supt. and Clerks Central Warehouse Overhead Insurance - Property " - Group, Health and Life " - Group Annuity	13,494.00 7,169.28 2,033.40 482.35 835.64 2,016.86
Personal Injury - Comp. & Doctors Taxes - Unemployment Insurance " - Old Age Benefit Electrical Engineering Department Employees Insurance & Compensation Design Department	440.37 348.00 42.54 401.04 396.64 1,260.00 16.91
Total General Mine Expenses Total Cost as Above	38,148.44 86,370.87
Proportion of Taxes	35,304.00
Depreciation - Movable & Misc. Equipment	829.02
Loading and Shipping	64.58
TOTAL IDLE EXPENSE	122.568.47

Mather "A" 1956

5. ESTIMATE AND ANALYSIS OF ORE RESERVES:

The net ore reserves reported to the Michigan Tax Commission on December 31, 1956 were 8,238,042 tons. This is a decrease of 1,839,833 net tons from the 1955 estimate. Included in the 1956 estimate are 863,222 net tons in Section 1, 47-27, to be mined by Mather Mine, "A" Shaft.

Reserves on the 8th Level and above were decreased and reserves between the 8th and 9th Levels increased. The increase in reserves between the 8th and 9th Levels was not sufficient to offset the decrease on the 8th Level and above.

Mining, a decrease in the anticipated reserves of some of the ore areas due to contamination by large intrusives, abandonment of small isolated areas for economic reasons, and a decrease in the size of some ore areas due to a more accurate delineation by development drifts and raises all contributed to the general decrease in reserves on the 8th Level and above.

The 1956 estimate indicates a net loss in reserves of 587,641 tons as compared to a gain in 1955 of 3,282,952 tons.

Estimated Net Reserves as of December 31, 1955	10,077,875
Production, January 1, 1956 to December 31, 1956	1,252,192
Net Reserves December 31, 1956 by Subtraction	8,825,683
Estimated Net Reserves as of December 31, 1956	8,238,042
Net Loss in Reserves	587,641

Expected Average Natural Analysis of Ore Reserves as of December 31, 1956

Grade	Tons	Iron	Phos.	Sil.	Mang.	Alum.	Lime	Mag.	Sul.	Loss	Moist.
Mather - Surface Diamond Drilling	551,938	53.15	.122	5.08	0.25	2.62	.58	.60	.014	1.97	12.50
Mather - Underground Development	7,686,104	51.62	.100	8.75	0.20	2.45	1.00	• 50	.050	2.25	11.00
	8,238,042										

The tonnage and analysis figures shown in the preceding table are the same as the figures that were turned into the Michigan State Tax Commission.

-12-

6. LABOR AND WAGES:

a. Employment:

The total mine payroll at the end of the year was 597 with a decrease of five men from the previous year.

Number of Men 1/1/56	602
Added to Roll During the Year	26
Total	628
Separations	31
Total on Payroll 12/31/56	597
Average Number of Men as per	
December Labor Statement	556

During the year there were thirty-one separations and twenty-six additions. Of the separations, twelve quit, three died, four retired, two were transferred to the Republic Mine, one was transferred to the Humboldt Mine, three were transferred to the Pelletizing Plant, one entered the service, two were disabled, one was discharged, and two were dropped from the payroll because of extended illness. Of the additions, two were transferred from the Lloyd Mine, five were returned servicemen, two were rehired, thirteen were Cliffs Shaft men, two were transferred from the Maas Mine, one was transferred from the Mather Mine, "B" Shaft, and one returned from a leave of absence.

b. Statement of Wages:

Avenage Wages Den Dav	<u>1956</u>	<u>1955</u>
Surface Underground Total	\$21.09 <u>23.97</u> \$23.40	\$18.15 <u>22.79</u> \$21.83
Average Wages Per Month Surface Underground Total	(213 Days) \$458.70 <u>521.35</u> \$508.95	(20 Days) \$363.00 <u>455.80</u> \$436.60
Tons Per Man Per Day Surface Underground Total	50.57 <u>12.53</u> 10.04	44.40 <u>11.56</u> 9.17
Labor Cost Per Ton Surface Underground Total	\$.417 <u>1.913</u> \$2.330	\$.410 <u>1.971</u> \$2.381

6. LABOR AND WAGES: (Continued)

c. Labor Relations:

Labor relations in general were satisfactory during the year.

One formal grievance was presented during the year. John T. Mattson submitted a grievance regarding seniority as applied to a job vacancy. His grievance was denied by management and later dropped due to the failure of the union to appeal the case to arbitration within the time limit specified in the basic agreement.

An industry wide strike affected the mine from July 2nd to August 6th. The strike was conducted in an orderly fashion. Salaried personnel and certain hourly rate personnel performed necessary maintenance work during the period of the strike.

The men benefited by six paid holidays, New Year's, Memorial Day, Independence Day, Labor Day, Thanksgiving, and Christmas, in accordance with the provisions of the labor contract.



7. SURFACE:

Buildings:

Only a minimum amount of maintenance was required on the mine buildings during the year.

Headframe and Trestles:

The headframe equipment operated throughout the year with only minor delays. The Kennedy Van Saun pan feeder was again overhauled with fifty new wheels, eighteen shafts, and seven new pads being installed. New wedge style pocket doors were installed for loading the trestle cars. Manganese wear plates were replaced on the rock pocket and the butterfly door. The railroad loading pockets and rock pocket section below the trestle deck was sheeted in to prevent snow and ice from building up in this area. Also closed in for improved winter operation was the Euclid truck loading area below the rock pocket chute. The north and south skip sheaves were ground to provide the correct operating clearances and the bearing blocks for these two sheaves were rebabbitted.

Stocking:

The trestle cars operated without any delays and only normal maintenance during the year.

Engine House:

The four 2700 c.f.m. air compressors were overhauled including the cleaning of electric motors and intercoolers. An electrical short in one of the skip hoist armatures caused an eight hour shut-down on April 17th. This was temporarily repaired and then changed with the spare unit later in the month. On April 19th the skip hoist was again shut down because of damage to the two drive pinions and bull gear. Operation resumed in eight hours after the raised surfaces of the teeth were filed smooth, and the gearing completly cleaned.

On November 25th the cage motor armature was removed for a thorough cleaning and replaced with a spare unit.

Skips:

The skips operated successfully throughout the year.

Hoist Ropes:

The cage rope was removed from service on September 2nd because of three loosened strands 1700-1900 feet from the cage end after operating three years, eight months, and eighteen days.

On May 26th the North skip rope was removed from service after hoisting 1,001,721 tons and on June 3rd the South skip rope was removed after hoisting 1,329,273 tons.

Mather "A" 1956

7. SURFACE: (Continued)

New Equipment:

A new fork lift truck for use in handling supplies on the property was purchased and the old concrete mixer for the batching plant was replaced with a new electrically driven unit.

Subsidence Studies:

D.D.H. #65 was re-entered early in February for the following reasons:

- (1) To attempt to locate the position of the void above the old 5th and 6th Level workings by extending the hole.
- (2) To follow progressively the advancement of the void towards surface.
- (3) To correlate all information so obtained with the general subsidence studies.

During the re-entry period, difficulties were encountered which caused abandonment of all attempts to deepen the hole.

Observations of significant value were:

- (1) There appears to be ground fracturing and movement at a depth of 1,000'.
- (2) After setting a cement plug below 800', the hole stopped taking air and water.

The hole was again cemented off at 767' and a microseismic recorder and other associated equipment were set up at the hole. A geophone was set in the hole and located at various depths until July 3rd. On July 30th, a newly designed geophone was set at the bottom of the hole, 767' below the collar, and remained there the rest of the year. The microseismic activity recorded at Hole #65 during the year indicated very little disturbance in the area under test. Enough activity was noted however, to warrant continued study in this area.

D.D.H. #68 was completed in May with the final effective bottom being at a depth of 861'. The collar of this hole is 100' West and 400' North of D.D.H. #65, and it was drilled at an inclination of -65° due South. The hole bottom is over the principal mined out area above the 5th Level.

In September, a geophone was installed at D.D.H. #68. It was set at hole bottom, 861' from the collar. During the remainder of the year, microseismic disturbances were present in the area but the disturbances were infrequent and of very low magnitude.

D.D.H. #69 was the third subsidence hole to be drilled in the Mather "A" area. It was collared 250' North and 400' East of D.D.H. #65, and drilled at an inclination of -65° due South. The hole bottom is over the principal mined out area above the 6th Level.

Mather "A" 1956

50

-16-

SURFACE: (Continued)

Subsidence Studies: (Continued)

The drilling was completed in September with a depth of 1,326' being attained. On October 31st, a geophone was lowered to a depth of 1,215'. It was readily discovered that the hole was not supporting a head of water and that air was flowing into the hole. Consequently, on November 7th a drill crew plugged and cemented off the hole at 1,150'. The seal is not entirely satisfactory but it does support a small column of water allowing the geophone to function properly.

Except for a group of noises recorded coincidently in Hole #65 and Hole #69 on December 26th, microseismic activity in the area was of very low magnitude. Holes #65, #68, and #69 are critically located and a triangulation network can be set up which might enable the performance of tests to establish the source of rock disturbances in the Mather Mine, "A" Shaft area above the 5th and 6th Level mining areas. For the present however, the plan to triangulate on the noise sources will have to be postponed until energy of greater magnitude is received by all geophones and a high percentage of the same disturbances recorded coincidently.

The subsidence studies undertaken during 1956 indicate that ground fracturing above the old 5th and 6th Level workings has reached an elevation between 1,200' and 1,000' from surface. However, the rate of subsidence as indicated by microseismic activity does not seem to be rapid. Further information of the rate of subsidence can be noted by determining when additional ground fracturing will cause the lowering of the water columns that are now present in D.D.H. #65, #68, and #69. Air entering or exhausting from the holes will give further indication that fracturing of the ground above the old workings has reached the effective bottoms of the subsidence holes.

During the month of September, Mr. L. Bacon, Geophysics Department, Michigan College of Mining and Technology, spent considerable time at the Mather Mine, "A" Shaft conducting a reflection seismograph survey. The results of the survey have not as yet been submitted. The purpose of the survey was to further attempt to locate the void over the old mining areas above the 5th and 6th Levels.

8. UNDERGROUND:

5th Level:

Production from the 5th Level was 26,490 tons, or 2.13% of the total mine production. All of the ore was Standard grade and came from over the #7 Cross-cut.

Drift development was completed and all the remaining reserves will be mined from the -160', -225' and the -275' elevations. It is anticipated that mining will be completed during the first half of 1957.

The crusher, which had been located over the #7 Cross-cut was removed and placed in service on the 9th Level. All 5th Level ore has since been crushed in the headframe.

7th Level:

Production from the 7th Level was 598,787 tons or 48.15% of the total mine output. The production came from seven areas in Section 2 and three areas in Section 1. The areas in Section 2 were as follows: North Block East, #5 South Cross-cut, #6 Crosscut, #1 Cross-cut, #5 North Cross-cut, #780 Block, and #7 South Cross-cut. The areas in Section 1 were the #7B Cross-cut, #8B Cross-cut and #9B Cross-cut.

During the year mining was completed over the #5 South, the #5 North and #6 Crosscuts. Reserves over the #5 North and #5 South Cross-cuts were exhausted, and heavy ground conditions over the #6 Cross-cut caused the end of operations in that area. The reserves remaining over the #6 Cross-cut will be recovered from the 8th Level. Of the four areas still active in Section 2, all but one are located in the pillar over the footwall drift.

Production for the year from the areas where mining was completed was as follows: #5 Cross-cut South - 2,983 tons; #6 Cross-cut - 49,775 tons; and #5 Cross-cut North - 45,000 tons.

Mining from the North Block East, which is located immediately above the loading end of the 7th Level belt conveyor, produced 56,255 tons. Two more mining transfers still remain to be developed in this area. 72,577 tons were produced from over the #1 Cross-cut. One transfer is all that will be required to mine the remaining pillar. The area over the #780 transfer drift produced 50,561 tons. Mining in this area is expected to continue for the greater part of 1957. 55,681 tons were produced from the #7 South Cross-cut and mining in this area, also, is expected to continue for the greater part of the coming year.

In Section 1, the #7B Cross-cut produced 81,551 tons; #8B Cross-cut - 55,428 tons; and #9B Cross-cut - 123,501 tons. All three cross-cuts are still active although no further development is required in the #7B and #8B areas and the remaining reserves over these cross-cuts are very small. The reserves in the area over the #9B Crosscut will permit mining throughout most of 1957.

Diamond drilling on the 7th Level consisted of two holes. U.H. #352 actually was drilled from the 6th Level on the 9962 W. section. The hole was drilled to test for ore northeast of present mining operations over the #9B Cross-cut on 7th Level. No enrichment was encountered.

Mather "A" 1956

62

-18-

8. UNDERGROUND: (Continued)

7th Level: (Continued)

In October an old diamond drill hole, U.H. #210, located along the 13,400 W. section, was re-entered after having previously been bottomed at 1,239'. The purpose of the hole is to test for ore above the footwall at 12th Level elevation. Drilling was still in progress at the end of the year.

8th Level:

8th Level production increased to 611,792 tons which represented 49.20% of the total mine production. The production came from five areas; the #7, #8, and #9 Crosscuts, and the #811 and #812 transfers. The latter two areas listed are loading transfers over the #81 sub-level belt conveyor.

Production from the various areas was as follows: #7 Cross-cut - 176,151 tons; #8 Cross-cut - 192,259 tons; #9 Cross-cut - 129,690 tons; #811 transfer - 55,861 tons; #812 transfer - 42,492 tons. A small tonnage was produced from development work for the #82 sub-level belt conveyor.

On the 8th Level, more than one-half of the reserves to be mined will be transported by belt conveyors. Two conveyors, which will service an area approximately 1,000' long along the strike of the ore body are required. These conveyors, which are discussed later, are called the #81 and #82 belts. The ore will be transported by the belts to an ore pass raise which will transfer the ore to the 9th Level. On the 9th Level the ore will be crushed and transported over the main 9th Level conveyor to the shaft.

During the year the #81 sub-level belt conveyor was placed in operation and has worked successfully to date. This belt has carried a total of 98,357 tons.

Development work for the #82 belt conveyor was completed and development for mining above the belt is well underway. Installation of the #82 belt conveyor will coincide with the completion of the 9th Level crushing and conveying system.

During the year four diamond drill holes were completed on the 8th Level. U.H.'s #315, #318, and #319 were drilled from the #3 Cross-cut to outline the ore body above the level in advance of development for mining. U.H. #358 was drilled from the footwall drift on the 10,075 W. section to outline the ore east of present mining operations in the #9 Cross-cut.

9th Level:

Work on the 9th Level consisted of main level drift development, cross-cut development, development of sub-level belt conveyor drifts, development for block caving over the sub-level belt conveyor drifts, completion of the excavation work for the main 9th Level belt conveyor system and installation of the main belt conveyor.

Three sub-level belt conveyors are planned for the 9th Level. These conveyors will handle practically all of the production from the level. The conveyors will work in progression with the #93 conveyor discharging onto the #92 conveyor, the #92 on-to the #91 and the #91 into the loading trench of the main 9th Level belt conveyor.

Mather "A" 1956

8. UNDERGROUND: (Continued)

9th Level: (Continued)

The drift for the #91 belt conveyor has been completed, the drift for the #92 conveyor is half completed, and work preparatory to development of the #93 conveyor was underway at the end of the year. Development for block caving over the #91 and #92 conveyor drifts is in progress. Installation of the #91 belt conveyor will coincide with the completion of the main 9th Level belt conveyor system.

A brief summary is shown below in tabular form for the various conveyor projects that are being developed or are installed.

		Conveyor Length	Belt Width	Belt Speed	Number of Loading Points
*Main Conveyor	- 7th Level	25001	30"	500 1/min.	l
*#81 Sub Conveyor	- 8th Level	400'	36"	350'/min.	2
#82 Sub Conveyor	- 8th Level	4801	36"	535'/min.	3
#91 Sub Conveyor	- 9th Level	2001	36"	525'/min.	1
#92 Sub Conveyor	- 9th Level	4901	36"	535'/min.	2
#93 Sub Conveyor	- 9th Level	6901	36"	535'/min.	3 or 4
*Main Conveyor	- 9th Level	3100!	30"	500'/min.	1

*Installed.

The 9th Level footwall drift was connected through to "B" Shaft during the month of December with excellent results on alignment and grade. Five levels, the 5th through the 9th, now connect "A" Shaft and "B" Shaft.

The following table indicates the amount of main level development completed during the year.

	Timbered Ore Drift	Timbered Rock Drift	Naked <u>Rock Drift</u> *	Total
NM 116 - 8th Level	Sall States	12 1 3 - 18 M	681	681
NM 116 - 9th Level	-	1,700'	130	1,830
NM 113 - 9th Level	195 - 196 - 196 March	1,163	110	1,273
	11. 11 C. 1	2,863'	3081	3,171'

*Rock bolted.

In addition to the projects listed above preparatory work was begun on a belt conveyor drift that will lift ore from the -2100' elevation to the 9th Level. Because of the length of time required to reach a depth three levels below the present operating level, and then to bring the new levels into production, it is important that the development is started at this time. The cage, ladder and pipe compartments of the present shaft will be deepened to handle men and supplies.

Eleven underground diamond drill holes, U.H. #314, U.H. #323, U.H. #324, U.H. #332, U.H. #333, U.H. #334, U.H. #335, U.H. #342, U.H. #343, U.H. #344, and U.H. #357, were drilled from the 9th Level during the year. All the holes except U.H. #332, U.H. #333,

Mather "A" 1956

64

-20-

8. UNDERGROUND: (Continued)

9th Level: (Continued)

and U.H. #334 were drilled to outline the ore body above the 9th Level. U.H. #332, U.H. #333, and U.H. #334 were drilled from a location south of a major East-West trending fault and along the 11,100 W. section. The purpose of these particular holes was to establish the dip of the East-West fault and to begin the exploration of the area south of the fault and below the 9th Level. U.H. #334 was drilled due south and at a dip parallel to the dip of the fault. A total of 155' of ore was encountered in this hole above the -2100' elevation. This ore was correlated with the ore encountered in Surface D.D.H. #44.

Mining Methods and Developments:

Block caving with the radial drilling system of undercutting continued as the mining method during the year.

Steel sets continued as the principal means of support in both main level and sublevel development. Four different type sets were used during the year. A threepiece set consisting of a 9' cap and 9' legs was used in main level development; a rigid arch set was used for the development of sub-level belt conveyor drifts, for cutting raises and for tugger rooms; a yielding arch set was used in undercutting drifts; and a three-piece set consisting of a 6' straight cap and $7\frac{1}{2}$ ' straight legs was used for transfer drifts.

Extensive use of 42" diameter steel raise tube liners was practiced during the past year. Use of the liners has reduced maintenance on raises considerably.

Mining Engineering:

Each month a complete survey of the mine was made to map the geology and the work completed in the various areas of the mine. This information was used to prepare four sets of maps which were distributed to the district superintendent of underground mines, mine superintendent, mine engineer, and mine captain. At the end of each quarter an additional set of maps was prepared and sent to the Bethlehem Steel Company. Ten sets of point maps were prepared each month for use by the supervisory personnel underground.

Routine survey work required the greater part of the time of two two-man survey crews. All development contracts were provided with lines and grades and frequently rechecked during the progress of the work. Usually approximately twenty crews were engaged in sub-level development work and two to three crews in main level development work.

Other work accomplished during the year was as follows: Ventilation survey and posting of fire maps, shaft gauging, tax estimates, stope analysis, stockpile analysis of iron, silica, and sulphur with accumulative totals for each pile being carried, stockpile surveys, accumulation of data on underground water, preparation of quarterly steel and timber requirements.

Weekly and monthly reports on all activities at the mine were prepared. A separate report of the activities most directly concerned with the mining engineering phase of the operation was also prepared each month.

Mather "A" 1956

65

-21-
UNDERGROUND: (Continued)

8.

Statement of Timbering	Supplies used in Op	erating Accounts
ITEM	AMOUNT	COST PER TON
Cribbing Stulls Lagging Poles Steel Minecrete Supplies	\$ 9,393.04 3,282.42 20,288.69 12,606.72 380,435.41 135.76	\$.00750 .00262 .01620 .01007 .30382 .00011
Total 1956	\$426,142.04	\$.34032
Total 1955	\$291,747.41	\$.2791

Explosives:

The following tables show the cost of explosives used in mining 1,252,192 tons of ore (Table I), the unit costs and consumption of explosives (Table II), and the cost per ton of explosives used in Development for Mining as compared to the cost per ton for mining (Table III).

TABLE I

Cost of Explosives - Operating

Powder - All Kinds Miscellaneous Blasting Supplies (Fuse, Caps, Bags, etc.)	<u>1956</u> \$100,849.90	\$ 93,865.71	
	54,050.11	43,711.95	
Total	\$154.900.01	\$137.577.66	

TABLE II

Unit Costs and Consumptions of Explosives

Pounds of Powder per Ton of Ore Tons of Ore per Pound of Powder Cost per Ton for Powder Cost per Ton for Fuse, Caps, etc. Cost per Ton for all Explosives	<u>1956</u> 0.443 2.256 \$0.081 \$0.043 \$0.124	<u>1955</u> 0.500 2.000 \$0.090 \$0.041 \$0.131
TABLE	<u>111</u>	
Cost per Ton in Development for Mining Cost per Ton in Mining	<u>1956</u> \$.042 .082	<u>1955</u> \$.040 <u>.091</u>
Total	\$.124	\$.131

Mather "A" 1956

66

-22-

8. UNDERGROUND: (Continued)

Pumping:

Pumping of underground water was continued from the 3rd and 6th Levels. The average total pumping rate decreased from 432 gallons per minute in 1955 to 425 gallons per minute in 1956, a decrease of 1.6%.

The average pumping rate from the 3rd Level decreased from 84 gallons per minute in 1955 to 59 gallons per minute in 1956. This is due to the fact that "A" Shaft no longer pumps the water from the 8th Level Cambria-Jackson workings.

The average pumping rate on the 6th Level increased from 348 gallons per minute in 1955 to 366 gallons per minute in 1956. The 6th Level pumps handle all the drainage below the 3rd Level, together with the 7th and 9th Level pump discharge, and all of the water from "B" Shaft. The "B" Shaft water increased from an average of 166 gallons per minute in 1955 to an average of 182.5 gallons per minute in 1956.

	Mather Mine, "A" Shaft Water G.P.M. Av.	Mather Mine, "B" Shaft Water G.P.M. Av.	Cambria-Jackson Water G.P.M. Av.	Total Water Pumped G.P.M. Av.
<u>1955</u> 3rd Level 6th Level Total	64 182	166	20 -	84 <u>348</u> 432
<u>1956</u> 3rd Level 6th Level Total	59 183.5	182.5	=	59 <u>366</u> 425

The mine discharge water is carried through a 16" pipe line and then by ditch to the Carp River, approximately two miles towards the west.

To provide for greater pumping capacities on the lower levels, various changes and additions have been made in the discharge columns and pump stations.

9th Level Pumping:

A 500 gallons per minute pump purchased from the Spies Mine was installed in the 9th Level pumphouse, thus increasing the pumping capacity to 1500 gallons per minute from the 9th to 6th Levels. To carry this additional water a 6" pipe line will be used.

6th Level Pumping:

A 500 gallons per minute automatic Ingersoll Rand pump was relocated in the 6th Level pumphouse. This pump handles all the 6th Level water under normal operating conditions and pumps to the 1st Level through a 6" column. A 400 gallons per minute Byron Jackson automatic centrifugal pump acts as a standby unit for the Ingersoll Rand pump.

Preparations have been made for the installation in the 4th pump stall and adjacent area of three automatic series pumps to handle 1,000 gallons per minute to surface. Delivery of these pumps is scheduled for January 1957 with the installation to follow

Mather "A" 1956

67

-23-

8. UNDERGROUND: (Continued)

6th Level Pumping: (Continued)

immediately thereafter. With the three 500 gallons per minute Worthington plunger pumps, the total pumping capacity will be 3,000 gallons per minute from this level.

New 10" Discharge Line:

A new 10" discharge line from the 3rd Level to surface was completed during the year. This allows the original 10" column to handle only 6th Level water while the new column carries 3rd and 1st Level water.

Alarm System:

An electrical alarm system will be used to give a warning in the event of failure of pumps on any level. Installation of this equipment was underway at the end of the year and when completed will assure efficient operation of the automatic pumping system.

9. TAXES:

Taxes for the year at the Mather Mine, "A" Shaft totaled \$342,338.52. The assessed valuation set for Section 2, 47-27, by Mr. Hardenberg, the state mine appraiser, was \$1,180,000 higher than in 1955.

		1956		1955
	VALUATION	TAXES	VALUATION	TAXES
Mather Mine "A" Shaft including Stockpiles, Supplies & Equipment as placed by State Mine Appraiser:				
Real Estate Personal Property Pipeline - Cloverdale Tract	\$7,440,000 950,000 650	\$303,552.00 38,760.00 26.52	\$3,795,000 3,415,000 <u>650</u>	\$155,595.00 140,015.00 26.65
Total Mather Mine "A" Shaft (Sec. 2, City of Ishpeming)	\$8,390,650	\$342,338.52	\$7,210,650	\$295,636.65
		N.		
	Press Lines	1956		
	TAXES	PER TON PRO	DUCED PER	TON SHIPPED
Total Operating	\$342,338.52	au. 213		₽0.500

and a state of the same		1955			
	TAXES	PER TON PRODUCED	PER TON SHIPPED		
Total Operating	\$295,636.65	\$0.283	\$0,201		

10. ACCIDENTS AND PERSONAL INJURY

There were 26 compensable injuries during the year with a lost time of 953 days for these injuries. There were 15 non-compensable injuries, which added 49 days lost time for a grand total of 1,002 days. The severity was 889 and the frequency was 36.36 compared with company averages for underground mines of 1,063 on severity and 38.60 on frequency. The total hours worked was 1,127,568 as compared with 1,030,395 in 1955 for an increase of 9.4%.

DATE	NAME	NATURE OF INJURY	DAYS LOST
1/18/56	Robert Lind	Multiple puncture wounds, forehead, face, neck. shoulders. chest.	7
1/18/56	Wiljo Leppanen	Multiple puncture wounds and lacerations face, neck, shoulders, chest, abdomen, arms. Fracture radius, mid, left.	99
2/17/56	Eino Leklin	Lumbo-sacral strain.	15
2/23/56	Sylvester Wiitala	Infected wound of finger extending into the palm of the hand. (Middle finger, right hand.)	25
2/22/56	Otto Talus	Two broken bones, right foot.	68
2/29/56	John Mattonen	Fracture, distal metacarpal, index finger, left.	27
2/29/56	Waino Lehtinen	Bruised left shoulder.	9
3/14/56	Michael Manzoline	Fracture 1st, 2nd, and 3rd metatarsal - right foot.	100
3/14/56	Richard LaMere	Contusion sacrum, left thigh. Strain lum- bar back.	9
6/ 1/56	Severino Guizzetti	Bad bruise on left thigh.	7
6/12/56	Arthur Seablom	Strain left inguinal region.	46
6/21/56	Louis Colombo	Lumbo-sacral strain.	25
6/24/56	Waino Mutka	Fractures 3rd and 4th fingers - left hand.	65
6/30/56	Salvatore Tasson	Contusion of both thighs. Sprained right ankle and fracture, right fifth metatarsal.	85
6/30/56	Norman Powers	Fracture of superior and inferior ramus of right ischial and pubic bones.	110
6/30/56	Henry Kuisti	Simple fracture of 8th rib axillary line, left side.	15
8/14/56	Charles E. Townsend	Muscle strain - lower back region.	10
8/22/56	Clarence Saari	Contusions, right posterior, hip region.	33
9/ 5/56	John Maki	Laceration, right hand - medial edge.	8
10/18/56	Maurice Hansen	Fracture middle phalanx, left index finger.	31
11/ 7/56	Walter Lummukka	Contusions due to crushing injury between timber truck and motor. Superficial abra- sions - no swelling. No fractured bones.	7
11/10/56	Dominic Carello	Fracture, right patella.	60
11/17/56	Arthur F. Eliassen	Fracture, middle phalanx, right hand.	50
11/23/56	Leslie Hutchens	Contusion left knee.	15
12/14/56	Leo McGlone	Contusion - severe left chest.	20
12/16/56	Vernon LaVeau	Cut little finger, left hand.	7
	and the second	Total Days Lost	953

Mather "A" 1956

20

NUMBER OF





11. POWER:

A total of 18,229,341 kilowatt hours of electric power was consumed during 1956. This was approximately 2,000,000 kilowatt hours greater than the preceding year, and is a direct result of the longer work week.

The power rate was determined by dividing the total operating cost of the Cleveland-Cliffs Electric Power Department by the total kilowatt hours sold and charging each consumer proportionately. To this is added a wheeling charge by the Upper Peninsula Power Company for distributing the power to the property.

	CONSUMPTION K.W. HOURS	AVERAGE MAX. DEMAND	AVERAGE DEM. FACTOR	COST OF CURRENT	AVERAGE PRICE PER K.W. HOUR
1956	18,229,341	3120 K.W.	68%	\$164,413.16	\$.0090
1955	16,245,161	3620	52%	150,266.76	.0092
1954	14,249,248	3550	46%	132,087.61	.0093
1953	17,431,000	3720	54%	283,853.29	.0164
1952	14,534,000	3120	54%	230,920.07	.0159

Mather "A" 1956

71

-27-

ASSELVED FLE CLEVE-CLAVE INDE CO SLEVELAND, DINO 72

MORRIS MINE ANNUAL REPORT YEAR 1956

57 JUN 12 AM 9:30

I. GENERAL:

The production for the year was 300,518 tons compared with 340,601 tons in 1955. The proportion of tonnage mined from Fee Lands was small due to operations approaching the 9th level elevation at the east end of the ore body and necessitating transfer of contracts to other areas. After the new development from the 10th level is completed into this area, additional contracts will be added and the proportion of production from this property will increase. The mine was idle five weeks in July and August due to the industry-wide strike.

A working schedule of 2-8 hour shifts per day for 5 days per week has been continued throughout the year. Hoisting is done on a 20-hour per day basis. A decrease in efficiency over the previous year is reflected in the comparison of 6.74 tons per man per day in 1956 with 7.00 tons per man per day in 1955. Total shipments were slightly in excess of production and amounted to 302,710 tons. The stockpile carry-over at the end of the year was 29,694 tons.

There was no exploration drilling done on surface or underground. A relatively small tonnage of new ore was proven by development and mining. The engineer's estimate of 3,013,308 tons shows a net gain of 63,460 tons in reserves after deducting the year's production from the 1955 estimate. The additional reserves were proven in Fee Lands and are the result of the 10th level development. A substantial reduction in reserves has been estimated in Chase Lease #24.

Deep-well surface pumping has been continued and an average of 1035 g.p.m. was pumped compared with 1299 g.p.m. in the previous year. Seven wells have been in operation but many of these have been very intermittently due to pump breakdowns and some delays due to power failure. The decrease in water pumped is due partly to the above reasons but chiefly because of a reduction in the output from Wells #8 and #10. A rapid decline took place during the year in the output from the above wells. The volume of underground water averaged 1600 g.p.m. compared with 1535 g.p.m. in 1955. The downward trend in underground water that occurred from 1952-1955 has been reversed with the increase that occurred in 1956.

Development of the 10th level has been underway throughout the year. A second sump drift was excavated on the west side of the plat to complete the development for the pump station. The first cross cut branching to the southeast from the main haulage drift was completed and stope development started above this drift. A footwall drift along the north side of the ore body was also completed as part of the development for the east end of the ore body. Some test drift was also advanced from the latter drift to outline the ore body more completely. All of this development has been in Fee Lands.

22. PRODUCTION, SHIPMENTS AND INVENTORIES:

a. Production

Year	Grade	Tons
1956	Morris	300,518
1955	1	340,601

MORRIS MINE ANNUAL REPORT YEAR 1956

2. PRODUCTION, SHIPMENTS AND INVENTORIES: (Cont^t d)

The 1956 production came from Fee and Leased Lands in the following proportions:

	Fee	Leased	Total
Production - Tons	54,009	246,509	300,518
Percentage	18.0%	82.0%	100.0%
Percentage - 1955	14.5%	85.5%	100.0%

A summary of the total production, Fee and Lease, since the Inland Steel Company took over the Morris Mine lease is listed below:

		Setting to the State		Tons	Percent
	Lease Ore Prod	uction 1933-19	56	5,455,054	75-4
	Tot	al		7,235,150	100.0
b.	Shipments			计同时上标语	
		Grade	Pocket	Stockpile	Total
	Contraction of the	Morris	167,470	135,240	302,710
	此时代, 11, 1	Grade	Fee	Lease	Total

Morris

The following table shows the shipments for the past five years:

50,494

252,216 302,710

Year	Total
1956	302,710
1955	335,939
1954	326,001
1953	324,150
1952	294,569
	the second state of the se

Total shipments since Inland acquired lease in 1933 - 7,195,891 tons.

c. Ore in Stock December 31, 1956

Grade	Grade		
Morris		29,694	

MORRIS	MINE
ANNUAL	REPORT
YEAR	1956
ALL STREET, ST	CARL COLORADO

74

2. PRODUCTION, SHIPMENTS AND INVENTORIES: (Cont'd)

d. Production by Months

Month	Days Worked	Average <u>No. of Men</u>	Tons Per Man Per Day	Production
January	21	197	6.65	26,688
February	21	196	6.65	26,738
March	22	196	6.29	25,892
April	21	194	7.26	28,296
May	22	195	7.22	31,570
June	21	200	7.29	28,788
July	0	0		225 (°)
August	20	196	7.07	26,214
September	19	197	6.98	26,582
October	23	198	6.83	29,422
November	21	196	5.71	21,057
December	_19_	196	5.46	19,936
Total	230	196	6.74	291,408
Stockpile Overrun				9,110
Total				300,518

(°) Carry-over from June

T. O.M.

e. Working Schedule

The mine operated 2-8 hour shifts per day 5 days per week.

f. Delays

There were no significant delays to operations during the year.

3. ANALYSIS:

a.	Shi	pmen	ts
	and the second s	the second s	

	Grade	Tons	Iron	Phos.	Silica	Mang.	Alum.	Sulphur	Moisture
	Dried Natural	302,110	55.34 49.01	•085 •075	13.70 12.13	•47 •42	2.65 2.35	.040 .035	11.44
b.	Ore in S	tock Dece	mber 31,	1956 (N	latural)				a and the second
	<u>Grade</u> Morris	Tons 29,694	<u>I ron</u> 49.36	Phos.	Silica 15.02	Mang. •43	Alum.	<u>Sulphur</u>	Moisture 11.00
c.	Ore Rese	rves - Ex	pected Na	tural An	alysis				
	Grade		Iron	Phos.	Silica	Mang.	Alume	Sulphur	Moistur
	Morris Hi-Sul	2,656,211 357,097	49.06	.071 .093	12.00 12.00	•44 •40	2.25 2.23	.015 .393	12.00 12.00

MORRIS MINE ANNUAL REPORT YEAR 1956

75

4. ESTIMATE OF ORE RESERVES

The estimated reserves, after allowance for ore mined in 1956, shows 63,460 tons of new ore developed. This tonnage has been proven between the 9th and 10th levels in Fee Lands as the result of 10th level development. In Chase Lease #24, a substantial reduction was estimated in reserves in Deposit #33 and also Deposit #79 due to major changes in the ore outline. In Chase Lease #9, an increase was estimated in Deposit #86 due to a revision in the ore area used for estimating purposes.

As development on the 10th level expands, particularly in a westerly direction, it is expected that a substantial tonnage of new ore will be added to the reserves in Fee Lands.

	Estimate 9-1-55	Production 9-1-55 to 9-1-56	Estimated Deducting Product	Actual Estimate <u>10-1-56</u>	Incr. or Decr. from 1955 Estimate
Chase Lease #24	95,853	576	95,277	39,342	55,935
Chase Lease #24 Hi-Sul	359,699	45,942	313,757	283,505	30,252
Chase Lease #9	1,972,471	177,253	1,795,218	1,803,175	7,957
Chase Lease #9 Hi-Sul	5,090	26,557	<u>21,467</u>	45,623	67,090
Total Chase Leases CCI Lands CCI Lands Hi-Sul	2,433,113 787,311 27,969	250,328 48,217	2,182,785 739,094 <u>27,969</u>	2,171,645 813,694 27,969	11,140 74,600
GRAND TOTAL	815,280	48,217	767,063	841,563	74,500
	3,248,393	298,545	2,949,848	3,013,308	63,460

5. LABOR AND WAGES

The labor force has remained practically constant as indicated by the average of 196 men in 1956 compared with 197 in the previous year. The industry-wide wage increase of 0.003 increase between job classes and $0.07\frac{1}{2}$ to base became effective July 1st, 1956.

6. SURFACE

The flume for the discharge from #9 surface well was rebuilt and also a new submersible pump installed in #3 surface well. Due to a likely probability of an extension of the Lloyd Mine surface cave involving the water main to the mine, a pumping plant was constructed near the Carp River, west of the mine, to furnish water for the mine.

To prevent bad icing conditions in the ventilation shaft, additional heaters were installed to pre-heat the intake air into the mine.

At the extreme west end of the 9th level the drift was dammed up for storage of water and a pump installed to deliver the water through a pipeline directly to the 9th level sump. This eliminated considerable mud and water from the haulage drifts and has improved tramming conditions.

Surface Pumping

The table below shows the amount of water pumped from surface wells in December 1955 and 1956:

MORRIS	S MINE
ANNUAL	REPORT
YEAR	1956
THE PERSON NUMBER OF THE PERSO	A RECEIPTION OF THE RECEIPTION

6. SURFACE (Cont'd)

Surface Pumping (Conttd)

	G.P.M.	G.P.M.
Vell No.	Dec. 1956	Dec. 1955
1	87	190
2	10	75
3	170	105
3A	257	306
5	El Altra - a a a	
8	125	275
9	92	83
10	272	318
	1013	1352

The average drop in the water level in the surface material above ledge, since pumping started in 1937, to December 28, 1956, is shown in the table below:

Test Hole	Drop 8-25-37 to 12-28-56	Depth Remaining To Ledge
501	79.7	13.8
506	63.4	23.8
510	36.9	86.7
511	41.1	113.0
514	30.3	97.6
515	19.9	108.7
517	29.3	82.8
522	30.0	83.5
524	17.7	65.1
527	51.9	23.1
528	12.2	87.3
531	5.8	70.9
534	1.4	95-4
Total	356.8	951.7
Average	27.5	73.2

Operating Expense for surface drainage amounted to \$16,997.00 compared with \$24,014.00 in 1955. The cost per ton was \$0.06 compared with \$0.07 in 1955.

7. UNDERGROUND

a. Pumping

The bulk of the underground water is being handled by the two automatically operated 1200 g.p.m. centrifugal pumps located on the 9th and 4th levels.

The following table shows a comparison of the mine water pumped over a five year period:

Year	4th	6th	7th	8th	9th	Total
1956	51.0	109.0	109.0	476.0	855.0	1600.0
1955	57.2	52.0	108.0	509.3	791.0	1534.6
1954	71.9	43.5	96.8	553.9	797.0	1574.2
1953	76.1	37.1	77.0	546.2	882.4	1620.9
1952	94.6	27.1	69.2	495.8	971.4	1658.1

MORRIS MINE ANNUAL REPORT YEAR 1956

12:2

7. UNDERGROUND (Cont'd)

a. Pumping

The following table shows a comparison of the underground pumping cost per ton for the last six years:

Year	Cost Per Ton
1956	\$.61
1955	.51
1954	•46
1953	•55
1952	.65
1951	•49

b. Development

The major development consisted of work on the 10th level. A suction sump and a pumphouse were excavated on the west side of the shaft plat. This development has been completed and is ready for the pump installation. Construction of the two scraping trenches north and south of the loading pockets was completed and also the incline drift from the 10th level to the skip pit.

The first cross cut branching to the southeast from the main haulage drift intersected the ore body and was stopped when the south footwall dike was reached. About 110° of ore was encountered in this development. Development for the first stope was started above this drift with the driving of a transfer drift east along the strike. About 500° of footwall drift was advanced to the east along the north side of the ore body. Small test drifts were advanced south from the drift at 200-foot intervals to determine the strike and outline of the ore body. A small test drift was also advanced due south from the end of the first cross cut for exploration purposes.

In Chase Lease #24, raise development was started from the 9th to 8th levels. This raise will serve as an ore pass or transfer for the production from #30 stope at the west end of the 8th level. The product from this area will be transferred to the 9th level for tramming to shaft and thereby eliminate the need to maintain tramming operations on the 8th level.

Chase Lease #24

An average of three contracts have mined in this Lease during the year. One contract continued stoping in Deposit #79 above the 8th level and development was advanced further east along the strike. At the close of the year, operations had crossed the east boundary of the Lease into Chase Lease #9 on the higher subs. At the west end of Deposit #79, caving operations have been continued and have approached very close to the 9th level elevation. In Deposit #82, a stoping operation was continued and operations completed about the middle of the year. Two contracts were conducting mining operations in this Lease at the close of the year.

Fee Lands

An average of four contracts have mined in Fee Lands and operations have been confined mostly between the 8th and 9th levels. A very small tonnage was mined at the 8th level and above by caving in Deposit #76 near the west boundary of the Lease.

In Deposit #87, a caving operation has been continued throughout the year and mining has extended across the east boundary of Chase Lease #9 into Fee Lands.



28

Constants Constants

es:em si lin is

7. UNDERGROUND (Cont'd)

b. Development (Cont'd)

Fee Lands (Cont'd)

At a lower elevation in Deposit #87, a stoping operation has been continued near the west boundary of the Lease. Development and mining, here, ranges from the 9th level to the -270' Sub. In Deposits #84B and #84C, at the extreme east end of the mine, caving operations have been continued and have practically reached the 9th level elevation. Relatively small pillars are all that remain to be recovered to complete operations in this area. Development from the 10th level will be driven into these deposits in the coming year so mining can continue below the 9th level.

Chase Lease #9

There was an average of ten contracts mining in this Lease during the year and the bulk of the production was from this property. Mining has been concen-trated in Deposits #33, 75C and 76, between the 8th and 9th levels. Sub-level caving has been employed mostly in these deposits with one sub-level stope operation being conducted in Deposit #33.

One contract has continued sub-level caving in Deposit #87 near the east boundary of the Lease and a considerable part of the operation here extended east across the line into Fee Lands as mentioned previously.

Near the central part of the 9th level in Deposit #86, a sub-level stope was developed and mining has continued here throughout the year. Development and mining here has progressed above a transfer drift near the 9th level to the 8th level and above.

Sub-level caving has continued to be used in most areas and sub-level stoping to a smaller extent.