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Experiments on a telluride ore and design of plant for treating same

Charles Le Clair King

John Severin Schroeder

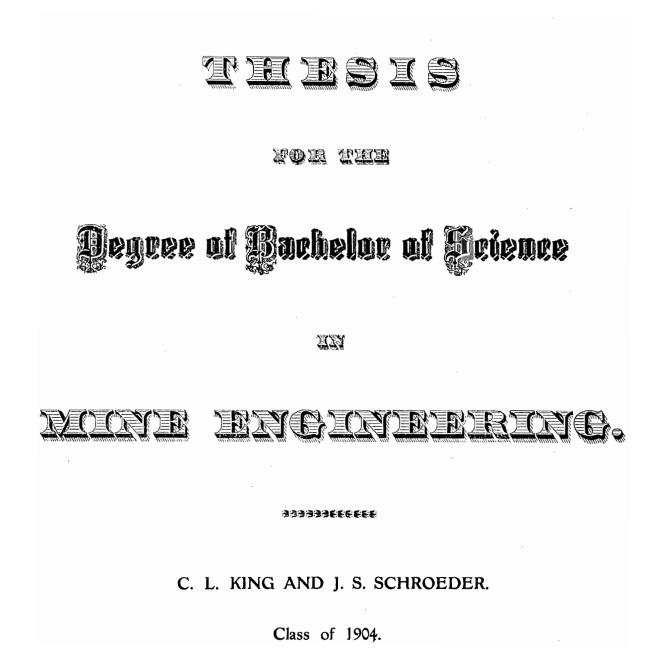
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Experiments on a Telluride Ore and Design of Plant for Treating Same.

The ore used in these pxperiments was a double telluride of gold and silver/coming fromLa Plata, Colorado.

The tellurides occur in thin seams of iron oxide deposited between two walls of very hard quartz. The quartz appeared tobe agatized in several of the samples. The veins range in thickness from one to two inches. The original state of the iron was in the form of either pyrite or marcasite which is plainly proved by the presence of pseudomorph crystals of these minerals.

About fifteen pounds of the ore was crushed down to ten mesh. It was well mixed and sampled down to one half. The sample was again well mixed and sampled down one half. This sample was crushed to forty mesh and sampled down to three quarters of a pound which was crushed to one hundred mesh and assayed by both wet and fire assay methods.

The fire assay was made by four methods:

By using an excess of Litharge. By using an excess of Soda. By nail method By secrification.

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The following charges were used in the different methods. Excess Litharge

> One half assay ton ore Thirty grams soda Ten grams borax Two and one half grams argol Forty grams litharge Salt cover

Results:

Au	3.82	ΟZ	Ag	.78
Au	3.80	0z	Ag	.80

Excess Soda

One half assay ton ore Thirty five grams soda Four grams argol Twenty grams litharge Salt cover

Result

Au 3.52 oz Ag .77

The check was rescorified with test lead before cupelling

Au 3.45 oz Ag .78

Nail Method

One half assay ton ore Thirty five grams soda Ten grams borax Three nails Twenty grams litharge Salt cover. Results

Au 3.74 ox Ag .51 oz

The check was rescorified

Au 3.73 oz Ag .47 oz

Scorification assay

One fifth assay ton ore Seventy grams test lead Ten grams litharge (used as a cover)

Borax

Results

 Au 3.42 oz
 Ag .64 oz

 Au 3.40 oz
 Ag .70 oz

The ore was then assayed by wet methods for gold, wilver and tellurium.

Results

Au 3.90 oz, #### Ag .80, Tel. 9.7

Au 3.90 oz Ag .78 Tel. 9.7

9.7 oz Fellurium equals .00033%

Laboratory tests made were chlorination, amalgamation and cyanide on both raw and roasted ore.

It was found that a combination of amalgamation and cyanide methods on the roasted ore gave the best extraction;

The amalgamation and cyanide tests were carried on in the following manner:

Some of the ore was crushed to forty mesh. About six assay tons were weighed into a mortar, enough water was added to make a thick mud. About twenty five grams of mercury were added and the mixture well stirred. The mercury was then panned out and the tailings saved. The mercury was strained but no amalgam was found. The tailings were then placed in a large funnel arranged with a stop cock below so as to regulate the flow of cyanide solution. 400 ee of a .5% cyanide solution was then poured on the tailings and allowed to percolate through very slowly. This operation was continued for twenty four hours. The tailings were then washed out and assayed.

Silver 2.94

Ag **G011 .57**

Giving an extraction of 25% gold and and 29% ag. The consumption of cyanide was .017%

Six assay tons of forty mesh ore were then taken and roasted beginning at a low temperature and finishing at a high temperature The roasting was carried on for an hour and a half. The lows in weight by roasting was about .4 grams

Roasted ore assayed:

Au 3.63 oz Ag .69 oz

Loss by roasting Au .17 oz, Ag .11 oz.

The roasted ore was then amalgamated in the same manner as the raw ore and the tailings saved. The amalgam obtained by straining was retorted and the residue wrapped in lead and cupelied.

Extraction was

Au 1.93 oz or 53%

Ag .37 oz or 53%

The tailings were treated with .5% cyanide solution for twenty four hours and assayed.

Au .34 oz Ag .08 Extraction Au 37% Ag 45% Consumption of cyanide was .006% Some of roasted forty mesh ore was amalgamated and treated with .1 .3 .5 and .7% solutions of cyanide and the tailings assayed.

Strength Solution	Assay Az	Value Ag	Percent Au	Extraction Ag	Consumption Cyanide
.1	1.54	.30	4	3	.001
.3	1.32 1.32	.24	11	10	.002
• 5	.34	.08	38	35	.006
.7	.34	.05	38	35	.009

The .5% solution gave the best extraction with the least consumption.

The next tests were made on sized 20/40/ and 60 mesh. Time twenty four hours. These were roasted/starting at a low temperature and finishing at a high temperature.

Five assay tons were weighed out and each one amalgamated and the tailings of each saved. The amalgam was retorted and cupelled. Extraction:

20 mes	sh Au 1.32 or	# 6%	Ag	.31	or	45%
40 me	sh Au 1.93 or	53%	Ag	.37	or	53%
60 mé	sh Au 1.98 or	54%	Ag	.39	or	56%
Tailings a:	ssayed					
20 mea	sh Au	2.31	Ag	.38		
40 me	sh Au	1.70	Ag	.32		
60 me	sh Au	1.65	Ag	30		

The tailings were then treated with a .5 % cyanide solution for twenty four hours. The tailings sere then carefully washed and assayed.

Size ore	Assay value		Percent	Extraction	Consumption
	Au	Ag	Au	Ag	Cyanide
20 mesh	.96	.28	37	10	.007
40 mesh	.34	.08	37	32	.008
60 mesh	.28	.05	38	33	.008

The next test was for time of leaching.

A roasted sixty mesh ore was amalgamated and some of the tailings treated with a .5% cyanide for twenty four, thirty six and forty eight hours. The tailings were then washed and assayed.

	Ass	ay value	Percent		Consumption
Time	Au	Ag	Extract Au Ag	ion	Cyanide .
24 hours	.28	.05	38 .3	3	008
36 hours	.12	trace	42 .4	4	009
48 hours	.07	5 6	44 .4		.01

The plant is to have a cap LOO T and will consist of a

Crushing Plant

Roasting Furnace

Amalgamation Mill

Cyanide Mill

The ore is brought direct from the mine and dumped on crusher floor. The crushing is done by a Comet crusher number "A" which crushes the ore to one and one half inch. The crushed ore falls direct into a storage bin. From this bin it is fed into a set of rolls which crushes it to three-quarters inch. After passing through these rolls it runs into the boot of an elevator which earries it up to a trommel. The undersize from the trommel passes direct into bin #2 below. The oversize passes through another set of rolls and then into bin #2. The final size being 1/4"

From bin number two it is conveyed by tramway to roaster building where ore is roasted in two Pearce Turret Frunaces. After being discharged from these furnaces it is wheeled of build ing to cooling floor.

When cooled it is loaded into car and hauled up an incline and emptied into bin number 3.

From bin number three it is conveyed by belt conveyor to stamp feeders. The stamps crush to sixty mesh and discharge into copper amalgamating plates. The tailings from these tables pass into a mercury trap and from there into settling tanks. When settling tanks are full water is drawn off and tailings loaded into cars and hauled to cyanide mill where they are placed in leaching vats. The leaching vats are filled to within a foot of the top and carefully leveled off. Great care being taken not to pack the tailings.

The track is carried on a trestle which is entirely separated from vats so that the vibration will not cause settling. The vats are built on a foundation separate and not connected with building The leaching solutions and wash water are introduced in the bottom of vat, causing them to flow up through the tailings.

The solution tanks of which there are two are placed in the highest level in the building about a foot above the leaching vats The solution from the leaching vats is run into a gold solution storage tank. There are two of these tanks, one for strong and one for weak gold solutions. From these storage tanks the solutions are run through the zinc precipitating boxes. The boxes have nine compartments 24" X 18" X 15". At intervals the precipitates are sluiced into a launder which conveys them to a settling tank. When enough are collected they are filted in a filter press and placed in an acid tank to dissolve excess of zinc. The residum are then melted down in a furnace. A vaccuum pump is used to aid leaching in vats and washing the tailings.

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