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TOMBSTONE MINING DISTRICT Cochise County, Arizona

Charleston Mine Area of Interest

Drilling
Reports, Notes, Assays &
Misc.

GEOLOGICAL

&

GEOCHEMICAL

STUDY

FOR: M. S. Horne

3033 North Central Avenue

Phoenix, Arizona

August 26, 1974

BY: C. A. Cosgrove, Mining Consultant

Working under supervision of: H. Clyde Davis, B.S; M.S.

Geologist & Mineral Director

(h)

LOCATION

The area covered under this study lies in the west 1/4 of Section 17 T 20 S R 22 E and forms a portion of the west slope of Uncle Sam Hill, about 3 miles southwest of Tombstone. There are 12 claims, 9 of which are in Section 17 and 3 of which are in Section 18. (See claim map appended to this report). These claims are listed as follows:

| Horne | #101 |
|-------|--------------|
| Horne | #102 |
| Horne | #103 |
| Horne | #104 |
| Horne | #105 |
| Horne | #106 |
| Horne | <i>#</i> 107 |
| Horne | #108 |
| Horne | #109 |
| Horne | #109-A |
| Horne | #159 |
| Horne | #160 |
| | |

GEOLOGY

This area is covered by an intrusive Latite Porphyry locally known as the Uncle Sam Porphyry. The principal outcrops of this porphyry are on the Uncle Sam Hill, the Three Brothers Hills to the north and west. Similar outcrops are also evident to the south near Charleston.

This porphyry is relatively resistant to erosion thus forming the craggy hills adjacent to the valley pediments. This porphyry has invaded the Bronco Volcanics, the Bisbee Formation, the Colina Limestone and the Epitaph Dolomite. This porphyry is probably younger than all the adjacent rocks except the Schieffelin Granodiorite and the Valley Fill of the San Pedro Trough. This intrusion is undoubtedly of the Tertiary period.

North of Uncle Sam Hill a deep embayment in the contact between the porphyry and the Bisbee formation, which forms the floor of the valley, suggests that the Bisbee underlies the porphyry at a comparatively shallow depth. This has been reported in the State of Maine mine, which penetrate



a considerable body of shale on the bottom level, at an elevation of about 4,300 feet (Butler, Wilson, and Rasor, 1938, p. 101). The floor at this part of the intrusion is therefore probably at a comparatively shallow depth, but farther southeast, near the Tombstone-Charleston road, the exposed contact becomes nearly vertical and cuts the bedding of the country rock at a high angle. Throughout the area between this road and the ridge extending north from Ajax Hill, the porphyry forms highly irregular masses that cut the Bisbee formation without regard to its bedding. A significant feature of this area is the mass of Uncle Sam porphyry following the major fault that farther north separates the Bisbee formation on the west from the Bolsa quartzite on the east. This fault was undoubtedly formed at the time of the major deformation of the Tombstone district. It had a minimum displacement of 5,000 feet and must extend to a considerable depth. The porphyry is frozen to both walls of this fault and has not been notably deformed since its emplacement, showing that the intrusion occurred after the major local orogeny.

A notable feature of the porphyry is the fracture (and strewing out of the resulting fragments) of a large proportion of the phenocrysts.

The groundmass ranges from vitric to microcrystalline, with a maximum grain size of about 0.04 millimeter, though more usually the groundmass crystals do not exceed 0.01 millimeter. Where determinable, the groundmass plagioclase is near Ango in composition and is contained, along with chlorite or biotite, in an intergrowth of orthoclase and quartz or glass.

Accessory minerals include magnetite, apatite, zircon, and sphene, but much of the sphene is a product of the alteration of biotite and presumable ilmenite. A few specimens contain rosettes of tourmaline, dichroic in greenish brown and brown. Some of this porphyry is mildly altered, with sericite, epidote, and albite developed in the plagioclase crystals and the mafic minerals altered to chlorite. Calcite is also present in some of the rocks.



CHEMICAL ANALYSIS OF UNCLE SAM PORPHYRY

| SiO ₂ | | | | 67.60 |
|--------------------------------|---|---|---|-------|
| 3102 | | | | |
| A1 ₂ 0 ₃ | | | | 16.22 |
| Fe ₂ 0 ₃ | | | | 2.01 |
| Fe0 | | | | 1.08 |
| MgO | | | | .91 |
| Ca0 | | , | | 2.91 |
| Na ₂ 0 | | | | 3.79 |
| к ₂ 0 | | | | 3.58 |
| H ₂ 0+ | | | | .92 |
| H ₂ O | | | | .28 |
| TiO ₂ | | | | .41 |
| P205 | | | | .19 |
| MnO | | | | .10 |
| ZrO ₂ | | | 8 | |
| o_2 | ¥ | | | |

The area under study has been cut by one minor fault or fracture. This bears north 40° east from the center of west line of Section 17. Radiating from this zone are numerous fractures which show filling by secondary mineral solutions - and considerable alteration to the porphyry in proximity to the zone.

A study was made of the high altitude photography by NASA to determine major structural factors effecting the possible implacement of crebearing solutions. A copy of an enlargement of this photography is attached herewith - This study was considered in the conclusion formed at the end of this report.



GEOCHEMICAL

Following the geological survey and preliminary surface study, it was determined that a geochemical reconnaissance would be the best practical method to locate anomalous mineralized areas for further study and drilling. Surface rock sampling was accomplished in accordance with the sample map attached herewith.

Geochemical results were determined by Skyline Labs, Inc., and are as follows:

| | Ag | Cu | Mo |
|-----------------------|------|-----------------------|-----|
| SAMPLE IDENTIFICATION | ppm | ppm | ppm |
| · | | | |
| 1A | <0.2 | 5 | |
| 2 3 | 0.2 | 5 | |
| | 1.0 | . 5 | |
| 4 | 0.6 | 5 · 5 5 5 | |
| 5 | <0.2 | 5 | |
| 6A | 0.2 | 5 | |
| 7B | 0.6 | 10 | |
| 8A | 0.2 | 5 | |
| 9 | 0.8 | 5 | 2 |
| 10 | 700. | 550 | 140 |
| 11 | 18. | 5 | |
| 12 | 1.4 | 5 5 5 | |
| 13 | <0.2 | 5 | |
| 14 | 0.2 | 5 | 2 |
| 15 | 1.8 | 5 | 2 |
| 16 | 0.4 | 5 | |
| 17 | <0.2 | 5 5 5 5 5 | |
| 18 | <0.2 | . 5 | |
| 19 | 0.4 | 5 | |
| 20 | 14. | 5 | |
| 21 | 4.2 | 5 5 | 12 |
| 22A | 2.0 | 5 | 12 |

Conclusions: It is evident that two anomalous areas exist which warrant detailed study. It is therefore recommended that an examination program consisting of drilling, testing and studies proceed to delineate mineralized areas. This examination program should be evaluated by feasibility reports as each step is concluded.

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This geological examination and research study together with the geochemical sampling was accomplished by C. A. Cosgrove, Mining Consultant, working under the direction and supervision of H. Clyde Davis, B.S; M.S., Geologist and Mineral Director for Brigham Young University.

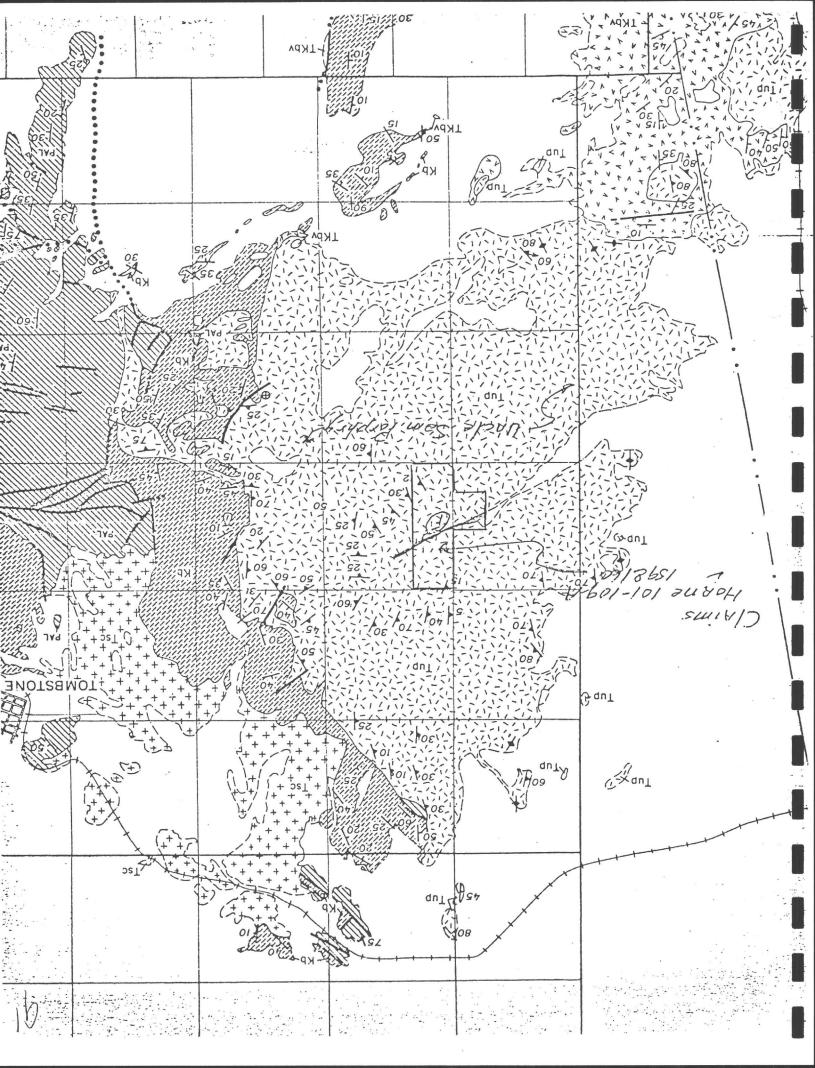
C. A. Cosgrove,

Mining Consultant

H. Clyde Davis, Geologist

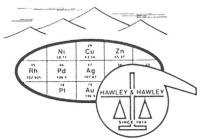
-5-

| | <u> </u> | 6h |
|---------------------------|--|-------------------------------------|
| | OMA HORNE # 101 | 40 |
| O ₁₆ A | @17A HORNE # 102 | |
| STATE MINTERAL LAND) TO | @15A @14A HORNE #103 @12 A | |
| SECTION 18 | OITA HORNE #104 | |
| HORNE # 108 10A | 09A 1-10RNE # 105 OZOA | SECTION 17. 7205 - R22E |
| HORNE #109 | 07B HORNE #106 | |
| HORNE # 109 A 5A @ | ⊕4A MORNE # (07 ⊕3A | |
| STATE MINERAL | HORNE #159 | |
| | HORNE# 160 | |
| | | SAMPIES-GEOCHEM 8-9-74 & 8-10-74 |
| | | |



| | | ENTIFIC | CATE O | F ANAL | YSIS | | Wil Arizona Re | liam L. Leh egistered Ass | mbeck |
|-----------------------------|-----------------|---------|---------|----------|------------------|------------|--------------------|------------------------------|-------|
| SAMPLE IDENTIFICATION | p pm | SILVER | COPPER | LEAD | ZINC | МО | Mn ppm | | 1 |
| Le Pepart - 159 8.160 acces | 0.12 2 mints | yw) | | | | | 156000 | | |
| Mr. C.A. Cosgrove | | | ARKS: | | CERTIFIED BY | / : | | | |
| 1923 West Osborn Road | | Tra | ce anal | ysis | - \$2.50 | | | | |
| Phoenix, Arizona 85016 | | i M | anganes | e - \$2. | 00 | | | HARGES: | |
| | | | | | | | | ALYSIS \$ | 4.50 |
| | | DATE | REC'D: | | COMPL.: /6/74 | | NUMBER: 41625-A | s | 4.50 |
| | | | | | | | | | |

There may be no area



SKYLINE LABS, INC.

Hawley & Hawley, Assayers and Chemists Division 1700 W. Grant Rd., P.O. Box 50106, Tucson, Arizona 85703 (602) 622-4836 Charles E. Thompson Arizona Registered Assayer No. 9427

William L. Lehmbeck Arizona Registered Assayer No. 9425

CERTIFICATE OF ANALYSIS

| ITEM NO. | SAMPLE IDENTIFICATION | Ag ppm | Cu ppm | Mo p.pm | | | | |
|---|---|---|-----------|------------|-----------|-----|-----|--|
| 1 2 3 4 5 6 7 8 9 10 | B - 1 2 3 4 5 6 7 8 9 | 2.6 1.2 0.2 <0.2 <0.2 <0.2 7.2 3.6 0.8 <0.2 7.4 | | | | | · · | |
| 11 12 13 14 15 | 11 12 13 14 15 | 2.8 <0.2 3.0 5.4 0.4 | | | | | | |
| 16 17 18 19 | B - 16 2 - 5 2 - 1-A 2 - H 102 E | 1.8 2.8 1.0 <0.2 | 25 5 (| 14 | | | | |
| | Justine Study of | - 100 139- | 109 R | | | | | |
| TO: | | | REMA | ARKS: | CERTIFIED | BY: | | |

| ro: Mr. C. A. Cosgrove | REMARKS: | CERTIFIED BY: | |
|---|--------------------|------------------------|--------------------|
| 1923 West Osborn Road Phoenix, Arlzona 65015 | Trace analysis | | |
| cc: Pacific Palisades, California | DATE REC'P: 9/4/74 | DATE COMPL. 9/10/74 | JOB NUMBER: 741777 |

James A. Briscoe & Associates, Inc.

Exploration Consultants:

Base and Precious Metals/Geologic and Land Studies/Regional and Detail Projects

James A. Briscoe Registered Professional Geologist Thomas E. Waldrip, Jr.

Geologist Landman

district the state of

January 2, 1987

Seth Horne, President JAMES STEWART COMPANY 707 Mayer Central Building 3033 N. Central Avenue Phoenix, AZ

RE: Letter report on geochemical samples taken from the south half of Section 36

Dear Seth:

We have finally received all of the assays back from the 35 samples that we took on the south one-half of Section 36, in order to satisfy the state prospecting permit work requirement for this year. We assayed these samples different elements including gold and silver of primary interest, and also copper, lead, zinc, molybdenum, arsenic, antimony, vanadium and mercury. We also intended to assay the samples for galium, germanium and uranium, however, the budget did not permit such analyses. In fact, through a mix-up on the assayers part, we ended up getting about \$1,200 worth of assays for \$800. When I submitted the samples, I presented him with your check for \$800, and asked him for the estimate for doing the afore mentioned assays. Instead of reporting to me, he simply started the process, and by the time he did give me a quotation, the work underway was approximately \$1,000. I explained to him that we did not have the budget for that. Since the analyses had almost been completed, he decided to go ahead and give us the results without further billing beyond our initial \$800.

The purpose for these multiple analyses was to determine whether the known metals associated with silver and gold could be used as path finders to the silver and gold mineralization.

The samples were taken from two parallel vein structures crossing your State land in the south half of Section 36. The host rock was in all cases Laramide age (approximately 65 million years ago) Bronco volcanics, consisting of andesite

Seth Horne, President JAMES STEWART COMPANY January 2, 1987 Page 2 of 4

laharic (mudflow) breccias. After deposition, these breccias have been cut by northeast trending andesite dikes, which are common throughout the western portion of the Tombstone Mining District, and northeasterly trending hydrothermal veins. The veins have been prospected by shallow bulldozer cuts in recent years, as well as small prospecting pits and shafts, possibly dating back before the turn of the century. A small tractor-mounted backhoe was used to trench through existing dumps and vein exposures, as described in the notes on Attachment 1.

Each sample is described on the notes portion of Attachment 1. In general, the sample consisted of from 10 to 15 pounds of rock material collected in a cloth sampling sack. In all cases, the rubber-tired backhoe was used to trench into bedrock or into existing dump material so that uncontaminated sample could be obtained. The trenches in bedrock were from two to approximately four feet deep. It is not too likely that gold values would have leached out of the surface, but it is conceivable that silver could be somewhat leached from the oxide zone and precipitated at greater depth at either the oxide sulfide interface or somewhere above that interface. The samples were processed in the Newmont Mining Company sample preparation lab. There, they were thoroughly crushed and pulverized to -10 mesh, thoroughly mixed, and then 200 grams split from the original sample. This sample was then ground to -300 mesh and submitted to the assayer -Copper State Analytical Lab, Inc. at 710 E. Evans, Tucson, Arizona. The gold and silver was assayed using the fire assay method with an AA finish. That is, the precious metal bead was obtained through the fire assay process, and then dissolved in acid and the amount of gold and silver determined very accurately using an atomic absorption spectrophotometer. The other elements were assayed appropriate analytical methods - for the most part, AA also. These described procedures were used to assure that the original sample was thoroughly mixed to insure a homogenous material before the assay sample was split out, and the assay would not allow any precious metal to go methodology undetected.

In the case of the dump samples, including 20 through 31, and the heap leach samples, including 32 through 35, trenches approximately 4 1/2 feet deep were cut through each dump. In the case of the small dump represented by samples 20 and 21, the samples were taken over approximately 15 foot intervals. In the larger dump, represented by samples 22 through 31, samples were taken over a 5 foot interval. The samples of

Seth Horne, President JAMES STEWART COMPANY January 2, 1987 Page 3 of 4

the heap leach - 32 through 35 - were taken from backhoe cuts at each corner of the pad, approximately 2 feet deep. These samples were felt to be representative of the dumps, and would show whether there was an erratic distribution of precious metals.

As can be seen by examining Attachment 1, samples 1 through 19, which represent everything except for the two larger dumps and the heap leach pads, there were no values even up to one part per million gold or more than 22 ppm silver - remembering that 34.285 ppm equals one troy ounce. Thus, it appears that all near-surface vein material along the two structures sampled contain only sub-economic amounts of gold and silver.

Samples from the larger dumps, where the old shafts made deeper penetration into the vein material, the results were also quite low. Sample 30, having 1.380 ppm (approximately 0.04 ounces) gold and 8.20 (0.24 ounces) silver, and sample #28 containing 0.280 ppm (approximately 0.01 ounces) gold and 22.20 ppm (0.65 ounces) silver, were the highest assays for gold and silver of all the samples taken. The maximum recovered gold and silver from these two samples would be about \$15.20 for sample #30, and \$4.88 for sample #28. The average recoverable precious metal content of the smaller dump was \$5.57, while that for the large dump was \$4.81, and \$3.76 recoverable gold and silver remained in the heap leach pad. Even if precious metal prices were to double, I don't see the circumstances that would allow these dumps to be worked at a profit.

CONCLUSIONS

The south half of State Section 36 contains two major vein structures which were sampled. The results are disappointing. Without much higher values than are indicated by the current sampling, the potential tonnages indicated are insufficient for profitable mining. The geologic data does not suggest any increase in width of the surface veins within 100 feet of the surface, and samples from the larger dump from the deeper shaft suggest that values within 100 feet of the surface are probably sub-economic.

RECOMMENDATIONS

The south half of Section 36 is comprised of Bronco andesite breccia for the most part. This breccia is propylitically altered except along vein structures where it is altered to

Seth Horne, President JAMES STEWART COMPANY January 2, 1987 Page 4 of 4

clay and sericite with silicification. Low values of gold, silver, copper, lead and zinc and other elements are present in these veins, though in sub-economic quantities. There is no geologic reason evident from the current study that would suggest the presence of an economically viable ore body in the south half of State the section. Therefore, it is recommended that no further money be expended in that area, and it be returned to the State.

The north half of the section, however, is comprised primarily or rhyolite dome material, and rhyolite ignimbrites, also of the Bronco series volcanics. They have not been sampled by this campaign. Because of the difference in rock character, the values from the andesitic terrain cannot be extrapolated into the rhyolitic terrain. Further, the Charleston Lead Mine, where alteration appears to be more intense, lies primarily in or adjacent to the rhyolitic terrain. Since assessment work for the northern half of the section must also be performed, it is recommended that a similar sampling campaign be done first in the Charleston Lead Mine open pit, by cutting fresh surfaces in the pit using the same backhoe, and then in surrounding prospects within the State section. If values are also sub-economic in the northern half of the section, then it is probable that the State land can be dropped from further consideration for a shallow, precious metal ore body.

Very truly yours,

James A. Briscoe

JAB/ms

Attachment

| | SAMPLE NUMBER | Au PPM | APM | Cu PPM | Pb PPM | 2r. PPM | Mo Piral | PFN | Sb PFM | V FFM | Ga D D M | Ge PPN | He n | U P P M | NOTES |
|---|------------------|-----------|-------|-----------|-----------|------------|-------------|-----|-----------|----------|-------------|----------------------------------|------|--|---|
| | 1 | 0.225 | 8.85 | 666 | 2.61 | 322.00 | 270 | | 11.00 | 64 | | | | | E of Lindsey Rd. & just E of Adobe building. Composite sample of dump de zer pile. Backhoe cut 15' long by 2/2 1/2' deep, FeOx veinlets in altered andesite. |
| | 2 | 0.045 | 2.20 | 172 | 0.28 | 106.00 | | | 1.50 | 28 | | | 83 | | 30' NE of sample #1. Sample of the S. end of 25' trench - 2 1/2' to 4' deep. Sample taken on 20 degree dipping vein with some FeOx & silicification in altered andesite. Sample from approx. 2' below surface |
| | 3 | 0.125 | 0.75 | 310 | 0.20 | 800.00 | 6 | 4 | 2.50 | 22 | | | 20 | | From N end of same trench - see sample #2 - in footwall of vein. |
| | 4 | 0.085 | 0.55 | 2.4 | 0.18 | 388.00 | 2 | 9 | -0.50 | 11 | | | 5 | one has the fire the set of the s | S of section line. Altered andesite with FeOx veinlets. No bedrock seen in sample cut to 2 1/2' deep. Composite sample taken from 1.5' down in 5' long trends. |
| | 5 | 0.185 | 1.50 | 26 | 240.00 | | 6 | | | 9 | | | <5 | | 10' E of #4, light colored altered material. Trench 10' in length by 3' deep. Verticle composite samples taken on each side of dump. |
| £ | 6 | 0.120 | 2.40 | 42 | 79.00 | 400.00 | 5 | 2 | 1.00 | 7 | | | <5 | | E-W cut in altered andesite bedrock. Trench is 5' in length l'below the surface of old dozer cut which was cut to approx. 5' below the surface. This is 300' E of #5 on trend of FeOx veinlets. Composite sample around wall of backhoe cut. |
| | 7 | 0.280 | 0.75 | 24 | 116.00 | 216.00 | 4 | 26 | 1.50 | 16 | | | <5 | | Samples 7 through 11 - these samples were taken in a N-S trench approx. 100' long by 5' to 6' deep. #7 in hanging wall. #8 in footwall. #9 in hanging wall, #10 in footwall near fault & veinlets, and #11 in silicified pods. All samples collected on west side of cut. |
| | 8 | 0.015 | 7.65 | 42 | 49.00 | 0.21 | 3 | 7 | 2.50 | 18 | | | <5 | | 100' long by 5' to 6' deep. #7 in |
| | 9 | 0.045 | 1.80 | 24 | 544.00 | 188.00 | 26 | 3 | 120.50 | 17 | | | <5 | | in hanging wall, #10 in footwall near fault & veinlets, and #11 in |
| | 10 | -0.005 | -0.50 | 9 | 20.00 | 426.00 | 3 | 21 | 120.50 | 12 | | | <5 | | silicified pods. All samples |
| | 11 | 0.085 | 0.85 | 28 | 164.00 | 0.19 | 14 | 74 | 1.50 | 13 | | | <5 | | |
| | 12 | 0.040 | 0.80 | 42 | 42.00 | 356.00 | 11 | 12 | 4.50 | 7 | | | 20 | | Samples 12 through 15 taken from a N-S trench, approx. 300' E of \$7 |
| - | 13 | 0.045 | 1.25 | 38 | 724.00 | 140.00 | 2 | 18 | 3.50 | 6 | | | <5 | | through #11, along trend of same vein zone. The trench is approx. |
| | 14 | 0.140 | 14.65 | 114 | 0.22 | 324.00 | 17 | 78 | 85.50 | 11 | | | 20 | | 75' long by 3' to 4' deep. Four samples taken; each one between 15' |
| | 15 | 0.015 | 3.60 | 160 | 0.47 | 00.008 | 7 | 19 | 3.00 | 7 | | | <5 | | Samples 12 through 15 taken from a N-S trench, approx. 300' E of #7 through #11, along trend of same vein zone. The trench is approx. 75' long by 3' to 4' deep. Four samples taken; each one between 15' à 20' long. They were taken across flat vein-like structures as seen in trench 300' to the west. #12 is at the S end & #15 is at the N end. #13 was from a flat vein with FeOx stain. Rock type in all cases is altered andesite. |
| | 16 | 0.035 | 3.40 | 84 | 0.16 | 308.00 | 1.5 | 21 | 14.50 | 8 | | 40 or 14 per per 11 or 12 per 14 | 15 | | Small trench 7' long, 75' further E of #15. Rock type is altered & FeOx veined andesite. |
| | 17 | 0.120 | 8.85 | 366 | 1.22 | 580.00 | 222 | 72 | 19.00 | 22 | × | | 85 | | On same structure, 300' to 400' E of #16. Composite sample taken around perimeter of NW trending cut, 3' deep by 20' long in old dump. |
| | 18 | 0.260 | 21.60 | 388 | 2.37 | 0.2% | 1.71 | | | 1.5 | | | 60 | | 25' NE of #17 along same structure. Trench dimensions & sampling methods were the same as above. |

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| LEMENTS UNITS | Au PPM | A P PPM | Cu PPM | Pb PPM | Zn PPM | Mo PPM | As PPM | St PPM | V PPM | Ga PPM | Ge PPM | Hg PPM | U PPM | NOTES |
|------------------|-----------|------------|-----------|-----------|-----------|-----------|-----------|-----------|----------|--|-----------|-----------|--------------|---|
| 19 | 0.035 | 1,15 | 28 | 0.16 | 99.00 | 6 | 2 | 2.50 | 11 | | | 65 | ************ | 400' NE of \$18. Sample from waste pile of old hand dug trench. Back-hoe cut 15' long, 1 1/2' deep on E side of trench. Composite sample around perimeter of cut. |
| 20 | 0.285 | 3.40 | 52 | 0.28 | 94.00 | 10 | 22 | 5.50 | 28 | | | 15 | | 600' SE of sample #19, sample #20 & #21 from old dump approx. 30' long |
| 21 | 0.185 | 3.85 | 66 | 1.02 | 100.00 | 16 | 11 | 8.50 | 284 | | | <5 | | \$ 10' to 15' wide. Sample #20 was from shaft to 15' SW. Sample #21 15' to 30' SW of shaft. Composite perimeter samples taken approx. 3' down in 5' deep backhoe cut. Dump |
| | | | | | | | | | | | | | | contains approx. 100 tons of material . |
| 22 | 0.110 | 2.80 | 106 | 0.44 | 444.00 | 68 | 52 | 4.00 | 22 | | | <5 | | 600' E of samples #20 & #21, sample #22 through #31 are taken from larg |
| 23 | 0.295 | 6.40 | 160 | 1.51 | 512.00 | 96 | 40 | 8.50 | 14 | | | <5 | | dump. Two trenches were cut at |
| 24 | 0.285 | 6.15 | 144 | 1.31 | 142.00 | 300 | 55 | 4.50 | 32 | | | 10 | | right angles to stratification in dump. The trenches form a narrow Y with arms trending to the SW. The |
| 25 | 0.315 | 8.40 | 166 | 1.88 | 126.0 | 550 | 110 | 8.50 | 18 | | | 15 | | or shorter leg of the Y was sampled at 6' intervals while the S long le |
| 26 | 0.300 | 10.65 | 134 | 2.11 | 122.00 | 560 | 2 | 6.00 | 26 | | | 35 | | of the Y was not sampled. Each of |
| 27 | 0.180 | 16.20 | 222 | 2.84 | 174.00 | 180 | 46 | 11.50 | 8 | | | 85 | | the composite perimeter samples was approx. 15 lbs. of material from a 6' deep trench, taken from about 3 |
| 28 | 0.280 | 22.20 | 800 | 6.70 | 324.00 | 460 | 2 | 13.5 | 12 | | | 145 | | in depth. The rock consisted of, |
| 29 | 0.340 | 8.45 | 148 | 1.98 | 134.00 | 450 | 45 | 11.00 | 11 | | | 60 | | for the most part, vuggy quartz, epithermal vein material with |
| 30 | 1.380 | 8.20 | 156 | 1.81 | 120.00 | 490 | 24 | 9.00 | 11 | | | 35 | | limonite after sulfides. Probably some sericite, though hard to see light colored vein material. The |
| 31 | 0.300 | 6.60 | 142 | 1.11 | 188.00 | 300 | 55 | 8.00 | 22 | | | 20 | | dump is composed of approx. 750 tor of material. |
| 32 | 0.440 | 5.85 | 156 | 1.04 | 306.00 | 324 | 21 | 4.50 | 22 | 100 May 200 AND 100 May 100 Ma | | <5 | | Samples #32 through #35 are taken |
| 33 | 0.235 | 5.40 | 180 | 1.35 | 204.00 | 470 | 42 | 2.50 | 76 | | | 15 | | on a heap leach pad dating back to circa 1980. This pad was very |
| 34 | 0.045 | 14.85 | 101 | 0.29 | 588.00 | 14 | 101 | 51.50 | 12 | | | <5 | | impermeable as indicated by ponded water. Because of its impermeabil |
| 35 | 0.160 | 49.40 | 256 | 0.36 | 0.17 | 17 | 109 | 95.00 | 10 | | | 20 | | ity, no production was probably ev attained. The heap material was |
| | | | | | | | | | | | | | | removed from the large mine dump sampled by #22 through #31. Sampl were taken from five 2' deep back hoe cuts at each corner & center o the heap. The heap itself was approx. 3 1/2' deep. Sample locations are: #32 SE corner, #33 center, #34 NW corner, & #35 NE corner. Approx. 500 tons of material are on the heap. |

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| SNAMBARARE | | n m n m m r : v | 3 5 1 1 4 C F 5 2 | # 2 2 A 3 11 5 2 2 2 | | 11 M P. L | | are a seco | 1 2 2 3 2 2 4 4 8 |
|------------------|-----------|-------------------------|------------------------|----------------------|-------------------------|--------------|-------|------------|-------------------|
| SAMPLE NUMBER | AU PPM | AU IN TROY OZ/TON | VALUE @ \$400 AU | AG PPM | AG IN TROY OZ/TON | \$6.00 AG | 90% | OF AG 6 | OF AU 5 AG |
| 20 | 0.29 | 0.01 | 3.32 | 3.40 | 0.10 | 0.59 | 2.99 | 0.30 | 3.29 |
| 21 | 0.19 | 0.01 | 2.16 | 3.85 | 0.11 | 0.67 | 1.94 | 0.34 | 2.28 |
| TOTAL | 0.47 | 0.01 | 5.48 | 7.25 | 0.21 | 1.27 | 4.93 | 0.63 | 5.57 |
| AVERAGE | 0.24 | 0.01 | 2.74 | 3.63 | 0.11 | 0.63 | 2.47 | 0.32 | 2.78 |
| 22 | 0.11 | 0.00 | 1.28 | 2.80 | 0.08 | 0.49 | 1.15 | 0.24 | 1.40 |
| 23 | 0.30 | 0.01 | 3.44 | 6.40 | 0.19 | 1.12 | 3,10 | 0.56 | 3.66 |
| 2 4 | 0.29 | 0.01 | 3.32 | 6.15 | 0.18 | 1.08 | 2.99 | 0.54 | 3.53 |
| 2 5 | 0.32 | 0.01 | 3.67 | 8.40 | 0.24 | 1.47 | 3.31 | 0.73 | 4.04 |
| 26 | 0.30 | 0.01 | 3.50 | 10.65 | 0.31 | 1.86 | 3.15 | 0.93 | 4.08 |
| 2.7 | 0.18 | 0.01 | 2.10 | 16.20 | 0.47 | 2.83 | 1.89 | 1.42 | 3.31 |
| 28 | 0.28 | 0.01 | 3.27 | 22.20 | 0.65 | 3.88 | 2.94 | 1.94 | 4.88 |
| 29 | 0.34 | 0.01 | 3.97 | 8.45 | 0.25 | 1.48 | 3.57 | 0.74 | 4.31 |
| 3 0 | 1.38 | 0.04 | 16.10 | 8.20 | 0.24 | 1.43 | 14.49 | 0.72 | 15.20 |
| 31 | 0.30 | 0.01 | 3.50 | 6.60 | 0.19 | 1.15 | 3.15 | 0.58 | 3.73 |
| TOTAL | 3.79 | 0.11 | 44.15 | 96.05 | 2.80 | 16.80 | 39.73 | 8.40 | 48.14 |
| AVERAGE | 0.38 | 0.01 | 4.41 | 9.61 | 0.28 | 1.68 | 3.97 | 0.84 | 4.81 |
| 3 2 | 0.44 | 0.01 | 5.13 | 5.85 | 0.17 | 1.02 | 4.62 | 0.51 | 5.13 |
| 3 3 | 0.24 | 0.01 | 2.74 | 5.40 | 0.16 | 0.94 | 2.47 | 0.47 | 2.94 |
| 3 4 | 0.05 | 0.00 | 0.52 | 14.85 | 0.43 | 2.60 | 0.47 | 1.30 | 1.77 |
| 3 5 | 0.16 | 0.00 | 1.87 | 40.40 | 1.18 | 7.07 | 1.68 | 3.53 | 5.21 |
| TOTAL | 0.88 | 0.03 | 10.26 | 66.50 | 1.94 | 11.63 | 9.24 | 5.82 | 15.06 |
| AVERAGE | 0.22 | 0.01 | 2.57 | 16.63 | 0.48 | 2.91 | 2.31 | 1,45 | 3.76 |

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COPPER STATE A NALYTICAL LAB, INC.

TUCSON, ARIZONA 85725

PH: (602) 884-5811

BILL TO-

James A. Briscoe & Associates 5701 East Glenn Suite 120 Tucson, Arizona 85712

| INVOICE: C | 6435 |
|-----------------------------|----------|
| JOB NO: 5360 DATE: 11/21 | |
| ACCOUNT NO: | |
| P.O.NO:Char | Mine |

COPY TO:

P A I D Check# 25045

| | ANALYTICAL CHAR | GES | | OTHER CHARG | ES |
|---|--|--|--|--------------------|----------|
| QUANTITY | DESCRIPTION | UNIT | AMOUNT | DESCRIPTION | AMOUNT |
| 35 35 35 35 35 35 37 7 | Gold Silver Copper Lead Zinc Arsenic Antimony Molybdenum Assays Mo 300 ppm | | \$201.25 \$ 91.00 \$ 70.00 \$ 63.00 \$ 56.00 \$ 157.50 \$ 166.25 \$ 91.00 \$ 43.75 \$ 14.00 | | |
| | AL CHARGES | marketing of a transfer of the second of the | \$953.75 | OTHER CHARGES | |
| to the of committee and more than committee and the | Less Profes Discou | | -\$153.75 | ANALYTICAL CHARGES | |
| | SOM THE PORT OF THE | 1-1-3 | \$800.00 | PAY THIS AMOUNT | \$800.00 |

NET 10 DAYS

707

LAB, INC. REGISTERED ASSAULK

LI HE .. H

710 E EVANS . TUCSON, AZ 85713

Place (COCH Blad Smit)



DESCRIPTION OF

James A. Briscoe & Associates 5701 East Glenn Street Suite 120 Tucson, Arizona 85712

Job: 5360 Received: 11/13/86 Reported: 11/19/86

Sample No 35 Elements: 9

Invoice No.- C 6435

| Elements | Αu | Ag | Cu | PЬ | Zn | As | | | |
|---------------|---|-------------|------|-------|------------|----------------|--|--|--|
| Units | PPM | PPM | PPM | PPM | PPM | PPM | | | |
| | | | | | ========== | | | | |
| I | 0.225 | 8.85 | 666 | 2.61% | 322 | 43 | | | |
| 2 | 0.045 | 2.2 | 172 | 0.28% | 106 | 2 | | | |
| 3 | 0.125 | 0.75 | 310 | 0.20% | 800 | 4 | | | |
| 4 | 0.085 | 0.55 | 24 | 0.18% | 388 | 9 | | | |
| 5 | 0.185 | 1.5 | 26 | 240 | 208 | 3 | | | |
| 6 | 0.12 | 2.4 | 42 | 79 | 400 | 2 | | | |
| 7 | 0.28 | 0.75 | 2.4 | 116 | 216 | 26 | | | |
| 13 | 0.015 | 7.65 | 4.2 | 49 | 0.21% | 7 | | | |
| K-2 | (), ()45 | 1.5 | 24 | 544 | 188 | 3 | | | |
| 10 | -().005 | -0.5 | 9 | 20 | 426 | 21 | | | |
| 1.2 | 0.085 | 0.85 | 28 | 164 | 0.19% | 74 | | | |
| 17 | (0.0)L | 0.3 | 4.2 | 4.2 | 356 | 12 | | | |
| | | 1. 25 | 38 | 704 | 140 | 18 | | | |
| 1 | 3 % | 1 12 . 17 6 | 114 | 0.22% | 324 | 78 | | | |
| 1.7 | 43 4 1 1 | 9 . r. | 160 | 0.47% | 800 | 1 () | | | |
| | | Y.4 | 87.6 | 0.16% | 398 | 2.1 | | | |
| - | | 1 23.5 | 366 | 1.22% | 5.8() | 7.2 | | | |
| g C | 1.70 | 21.6 | 388 | 2.37% | 0.24% | (j, \bar{c}) | | | |
| 1.5 | 0.035 | 1.19 | 2) (| 0.16% | 9.9 | 2 | | | |
| 30 | 0.295 | 3.4 . | 5.2 | 0.28% | Q /4 | 2.2 | | | |
| 2.1 | 0.145 | 0.85 | 66 | 1.02% | 100 | 7.1 | | | |
| 2.2 | 1. 1. | 2.1 | 106 | 0.44% | 444 | 5.2 | | | |
| 45 | 1 1 1 1 | 6.4 | 160 | 1.51% | 312 | 5 () | | | |
| 3.4 | 13 1 5 5 | 6.15 | 144 | 1.317 | 142 | r 5 | | | |
| | * | 8.4 | 1.66 | 1.197 | 126 | 110 | | | |
| | 0.3 | 10.65 | 134 | 2.11 | 122 | 2 | | | |
| Y 4. " | 0.18 | 16.2 | 222 | 2.84% | 1.74 | 46 | | | |
| 7 N | 0.28 | 22.2 | 800 | 6.7% | 324 | 2 | | | |
| * 1 1 | 0.34 | 8.45 | 148 | 1.98% | 134 | 45 | | | |
| 30 | 1.38 | 8.2 | 1.56 | 1.81% | 120 | 24 | | | |

LAB. INC.

710 E EVANS . TUCSON AZ 85713



James A. Briscoe & Associates 5701 East Glenn Street Suite 120 Tucson, Arizona 85712

Job: 5360

Received: 11/13/86 Reported: 11/19/86

Sample No 35 Elements: 9

Invoice No.- C 6435

| Au PPM | Ag PPM | Cu PPM | Pb PPM | Zn PPM | As PPM |
|-----------|---|---|---|--|--|
| | 12 12 12 12 12 12 14 15 15 15 15 15 15 15 15 15 15 15 15 15 | THE R. S. LEWIS CO., LANS. LANS. CO., LANS. | | | |
| 0.3 | 6.6 | 142 | 1.11% | 188 | 55 |
| 0.44 | 5.85 | 156 | 1.04% | 306 | 21 |
| 0.235 | 5.4 | 180 | 1.35% | 204 | 42 |
| 0.045 | 14.85 | 101 | 0.29% | 588 | 101 |
| 0.16 | 49.4 | 256 | 0.36% | 0.17% | 109 |
| | Au PPM 0.3 0.44 0.235 | Au Ag PPM PPM 0.3 6.6 0.44 5.85 0.235 5.4 | Au Ag Cu PPM PPM PPM 0.3 6.6 142 0.44 5.85 156 0.235 5.4 180 0.045 14.85 101 | Au Ag Cu Pb PPM PPM PPM PPM 0.3 6.6 142 1.11% 0.44 5.85 156 1.04% 0.235 5.4 180 1.35% 0.045 14.85 101 0.29% | Au Ag Cu Pb Zn PPM PPM PPM PPM PPM 0.3 6.6 142 1.11% 188 0.44 5.85 156 1.04% 306 0.235 5.4 180 1.35% 204 0.045 14.85 101 0.29% 588 |

| more that year plant street street them to the black proof order order at the street street. The street str | | |
|--|-----|------------|
| Elements | Sb | Мо |
| Units | PPM | PPG |
| | | |
| 31 | 3 | 300 |
| 3 2 | 4.5 | 324 |
| 33 | 2.5 | 474) |
| * Z | | 1 M 1 M |
| | 4.5 | 1 7 |



LAB, INC.

D.A SHAF

A. Z. William H. Hanne

710 E EVANS . TUCSON, AZ 85713

Ph 1607 284 581

James A. Briscoe & Associates 5701 East Glenn Street Suite 120 Tucson, Arizona 85712 Job: 5360 Received: 11/13/86 Reported: 11/19/86

Sample No 35 Elements: 9

Invoice No.- C 6435

| Elements Units | Sb PPM | Mo PPM | |
|-------------------|-----------|-----------|-------|
| | | | ===== |
| 1 | 1.1 | 270 | |
| 2 | 1.5 | 10 | |
| 3 | 2.5 | 6 | |
| Z _i | -0.5 | 2 | |
| 5 | -0.5 | 6 | |
| 6 | 1 | 5 | |
| 7 | 1.5 | 4 | |
| 4.2 | 2.5 | 3 | |
| 74 | 4.5 | 26 | |
| 1 7 | 120.5 | .3 | |
| l 1 | 1.5 | 1 4 | |
| 12 | 4.5 | 11 | |
| L"1 | 3.5 | 2 | |
| 1.7 | 85.5 | 1 7 | |
| | 2 | 7 | |
| | 14.5 | 1.2 | |
| | (·) | 222 | |
| | 38 | 176 | |
| | 2.5 | 6 | |
| | 5.5 | 10 | |
| 1 1 | 8.5 | 16 | |
| 7.2 | <u>7.</u> | 68 | |
| 113 | 8.5 | 96 | |
| | 4.5 | 300 | |
| | 8.5 | 550 | |
| | Ó | 560 | |
| | 11.5 | 180 | |
| , i. | 13.5 | 460 | |
| 20 | 11 | 450 | |
| 200 | Q | 490 | |

LAB, INC.

D.A. SHAH

AZ RIL # HHHH

710 E. EVANS . TUCSON, AZ 85713

FH 16021 884 5811



James A. Briscoe 5701 East Glenn Suite 120 Tucson, Arizona 85712 Job: 5360 Received: 12/02/86 Reported: 12/05/86

Sample No 35 Elements: 2

| Elements | Нд | V | |
|----------|-----|-----|--|
| Units | PPB | PPM | |
| | | | |

| Units | PPB | PPM | |
|----------------|-----|-----|---|
| | | | |
| 1 | 85 | 64 | |
| 2 | 8.5 | 28 | |
| 3 | 20 | 22 | |
| 4 | 5 | 1 1 | |
| 5 | -5 | 9 | |
| 6 | -5 | 7 | |
| 7 | -5 | 16 | |
| 8 | -5 | 18 | |
| 9 | -5 | 17 | |
| 1 () | -5 | 12 | |
| 1 1 | -5 | 13 | |
| 1.2 | 20 | 7 | |
| 13 | -5 | 6 | , |
| 1.4 | 2.0 | 11 | |
| 1.5 | - 5 | 7 | |
| 1.6 | 15 | 8 | |
| 1. 7 | 8.5 | 12 | |
| 18 | 60 | 18 | |
| 19 | 6.5 | 11 | |
| 20 | 1.5 | 28 | |
| 2.1 | 5 | 284 | |
| 22 * | - 5 | 22 | |
| 23 | -5 | 14 | |
| .) ⟨4 | 1() | 32 | |
| 25 | 15 | 18 | |
| 2.6 | 35 | 26 | |
| 27 | 8.5 | 8 | |
| $\frac{1}{28}$ | 145 | 12 | |
| 29 | 60 | 11 | |
| 30 | 35 | 1 1 | |

LAB. INC.

REGISTERED ASSAYER

D.A. SHAH

710 E EVANS . TUCSON, AZ 85713

PH (602) 884 581

James A. Briscoe 5701 East Glenn Suite 120 Tucson, Arizona 85712 Job: 5360
Received: 12/02/86
Reported: 12/05/86
Sample No 35
Elements: 2

Page 2

PERSONAL STREET

| \$100 time time time time time time time time | | | |
|---|---|-----|---|
| Elements | Нg | V | |
| Units | PPB | PPM | |
| \$100 Pages State S | ## No. 250 -7-2 Tag 100 -700 and age 200 And 500 -710 # 100 Dec 110 Mar 100 Pro 110 Bank 100 Mar 100 | | We will not |
| 31 | 20 | 22 | |
| 3.2 | - 5 | 22 | |
| 33 | 15 | 76 | |
| 3.4 | 5 | 12 | |
| 3.5 | 20 | 1.0 | |

PROPERTY OF THE SECOND SECOND

, Sto

October 15, 1971 Mr. H. Clyde Davis Director, Mineral Development 859 East 2730 North Provo, Utah 84601 Dear Clyde: We have sent you under separate cover via Airmail three small samples from Drill Hole #7 at Tombstone. We are still drilling one shift only, and make about 20 feet a day. Clarence and I think that you should have a serious study made of the samples enclosed yesterday. Note the unusual metal at approximately 3273'. Also you will note that we have run through a section of very good sericite. This is by far the deepest that we have encountered any sericite of this quality. Both Clarence and I will be in Washington, D. C. all of next week. We will be staying at the Interstate Inn (703) 534-9100. If you think that the hole should be shut down, please call us there and we will call Clark. I think Clarence has a surge of optimism because of what we have encountered at 3200'. Sincerely yours, M. S. Horne MSH:ef



784

CHEMICAL & MINERALOGICAL SERVICES 3435 SOUTH STATE STREET, SALT LAKE CITY, UTAH 84115 (801) 266-8228

ANALYTICAL REPORT FOR:

| Er. Clide Davis Brighem Young University | OUR NUMBER 2263 |
|--|-------------------------|
| Administration Bldg. | DATE Oct 24, 1971 |
| Provo, Utah Hole #7 | CUSTOMER'S ORDER NO. |

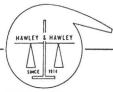
| Sample 🛉 | Ag ppa | Cu ppa | Po ppm | |
|----------|--------|--------|--------|---------------------|
| 3233 St | 3.0 | 20 | 10 | Light Rock - pyrite |
| 3274 St | k | 20 | | Darker Rock - |

03

.002

.001

Ray Brokerad



Registered Assayers OVER 50 YEARS

HAWLEY & HAWLEY

ASSAYERS AND CHEMISTS, INC. BOX 50106 1700 W. GRANT RD.,

TUCSON, ARIZONA 85703

(602) 622-4836



Douglas Hayden Morenci Inspiration El Paso St. Louis

Preparation \$ 15.30

Analysis \$

PNX 345569

161.50

176.80

| | | IDENTIFICATION | Gold opt | Silver | Lead % | Copper % | Zinc % | Mo. % | |
|------------|------------|-----------------------|-------------|--------|-----------|-------------|-----------|----------|-----------------|
| ., i il | <u>Ho1</u> | e 7 | | | | | | | |
| | 1 | 1656-1660 | | None | < 0.01 | < 0.01 | 0.02 | | |
| | 2 | 1670-1674 | | 0.01 | < 0.01 | < 0.01 | < 0.01 | | |
| | 3 | 1674-1678 | | None | < 0.01 | < 0.01 | < 0.01 | | |
| . | 4 | 1683-1687 | | None | < 0.01 | < 0.01 | < 0.01 | | |
| | 5 | 1697-1702 | | 0.01 | < 0.01 | < 0.01 | 0.14 | | |
| | 6 | 1702-1706 | | None | 0.03 | < 0.01 | 0.03 | | |
| | 7 | 1716-1720 | | None | 0.02 | 0.01 | 0.26 | | |
| | 8 | 1722-1725 | | 0.01 | 0.18 | < 0.01 | 0.01 | | |
| | 9 | 1775-1779 | | None | 0.02 | < 0.01 | < 0.01 | | |
| | 10 | 1839-1843 | | None | < 0.01 | < 0.01 | < 0.01 | * | |
| | 11 | 1885-1889 | | None | < 0.01 | < 0.01 | < 0.01 | | |
| " | 12 | 2102-2106 | | None | < 0.01 | 0.01 | < 0.01 | | |
| | 13 | 2161-2165 | | None | < 0.01 | < 0.01 | < 0.01 | | |
| | 14 | 2209-2213 | | None | < 0.01 | < 0.01 | < 0.01 | | |
| | 15 | 2218-2222 | | 0.01 | 0.18 | < 0.01 | < 0.01 | | |
| | 16 | 2231-2235 | | None | < 0.01 | < 0101 | < 0.01 | | |
| | 17 | 2236-2240 | | None | < 0.01 | < 0.01 | < 0.01 | | |
| | | | | | | | | | |
| | | | | | | | | | |
| | | | | | | | | | |
| 1 | | | | | | | | | 1 |
| | | | | | | | | | SEGISTERED ASC |
| | | | | | | | | | CERTIFICATE NO. |
| | | | | | | | | | (A AT ET WY |
| i. | | | | | | | | \\ | XIVITO II |
| | CC: | James Stewart Co. | | RE | MARKS: | | Analysis | Cert. By | Whed / |
| | ADD: | Attn: Mr. Clark Hu | | Si | ngle an | alyses | | | Rona U. S. A. |
| | CITY: | 3033 N. Central Avenu | | | 5 | , | | | |

Date Spl. Received 8/20/71 Compl. 8/25/71

....

'DD:

ACC:

TY:

Phoenix, Arizona 85012

JAMES STEWART COMPANY

September 19, 1973 MEMO TO: M. S. Horne RE: HOLE #8 It is my belief from observation that transition from volcanic to sedimentary rock occurred at a depth of 840-845 feet. At this point there is approximately six feet of fault gouge very, very heavy in ground up metals. No possibility of determination of type without microscopic examination and analysis. Thorough lab studies and assays of this hole would be to your advantage. A piece of core from 870' location is attached. CAC/bde C. A. Cosgrove The state of the s

September 19, 1973 MEMO TO: M. S. Horne RE: CHARLESTON Hole #8 was stopped at depth 872', Monday, September 17. Strata coordination was found between sediments: -- 735-797 850-864 A quartzitic material, one which I call sugar quartz. I think Clyde knows what I mean, and can give it its proper nomenclature. Below this strata there is a transition into fine grained sandstone, very salicaous. It is well to remember in comparing strata differentials that the collar elevation of Hole #8 is about 50 feet below collar elevation of Hole #2. Hence, the true strata differential is about 115 feet in a horizontal of 340' or a pitch of an angle whose tangent is .0334 plus or minus. CAC/bde C. A. Cosgrove

Hole #8

ARIZONA TESTING ABORATORIES

A DIVISION OF CLAUDE E. McLEAN & SON LABORATORIES, INC. 817 WEST MADISON ST. PHOENIX, ARIZONA 85007

PHONE 254-6181

For Mr. C. A. Cosgrove

Date September 5, 1973

Sample of ⋈ێ₩₩ Ore

Received: 9/4/73

Submitted by:

same

ASSAY CERTIFICATE

Gold figured at \$ 100.00 per ounce

Silver figured at \$ 2.00

per ounce

| LAB. NO. IDENTIFICATION GOLD SILVER OZ.PERTON VALUE OZ.PERTON VALUE | PERCENTAGES COPPER LEAD | ZINC |
|---|----------------------------------|-----------------|
| OZ.PERTON VALUE OZ.PERTON VALUE | COPPER LEAD | ZINC |
| | | 7 |
| S152 | 15ppm 35ppm 35ppm 5.00% 70 0.37% | 0.59% 1000pp |
| RECEIVED | | |

JAMES STEWART CO. Phoenix, Arizona

Respectfully submitted,

ARIZONA TESTING LABORATORIES.

ISEP 1 2 1973

Claude E. McLean, Jr.

Plande E.M E Lean fr.

September 19, 1973

MEMO TO: M. S. HORNE

RE: ASSAY ON LAST SAMPLES

Hole #8, Depth 771: Silver .2

Lead 0

Copper .025

Zinc 2.4

Hole #8, Sample #2, Depth 806: Silver .09

Lead 0

Copper .025

Zinc 0

all was pyrite.

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2

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an a

JAB

October 10, 1973

Mr. H. Clyde Davis 859 E. 2730 North Provo, Utah 84601

Dear Clyde:

RE: CHARLESTON - HOLE #9

Under separate cover we are transmitting samples of the volcanics from Hole #9. A brief study of the hole Monday and Tuesday indicated roughly the following.

The first 170 feet is very good serracite completely altered. We then have a horst of andesite, followed by another 40 or 50 feet of serracite. We then break into the andesite which continues down to approximately 515 feet. The top 100 feet of this andesite are fairly heavily metalized with pyrites diminishing as it approaches the 320 foot zone. At 340 feet we pick up a section which I believe is a minable zinc ore section flowing out from both sides of a high grade streak at 346 feet. This high grade streak has zinc sulphide, lead and calcopyrite. The calco extends a couple of feet in each direction from the center, but is the minor constituent of the three metals, with zinc predominant. I have not disturbed this ore zone, but have taken a sample from approximately 15 feet below it and having it assayed by Holley & Holley at Tucson. We are assaying this for all the metals.

The zone immediately below this ore zone for the next 150 feet is interspersed continuously with seams of sphalerite ranging from knife thickness to 1' thick zones. In almost every case the outer fringes of the sphalerite is seemingly oxidized into what I feel is hemimorphite. Also continuously through this section are isolated splotches of possible hemimorphite, some of them with the zinc as a pseudo morph after galena. Continuing on down, we pass the 520 foot zone and go into a couple hundred feet of highly salicious light colored volcanic monzonite heavily metalized with pyrite. However, much of the pyrite is oxidized and this zone also contains the green splotches. There falls below this a zone of a darker material, less heavily metalized with pyrite, but much heavier in the dark green splotches. Also this zone contains intermittent zones of the sphalerite, both in seams and in pocket forms. This hole bottoms out at approximately 900 feet deep with a 30 foot section of breccia.

gle

Mr. H. Clyde Davis

-2-

October 10, 1973

They were still in the braccia when I left last night and I presume they will have penetrated the sediment today. It is our intention to cut off the hole today and await study of both Holas #8 and #9 before further work is done. The holes are cased for further depth drilling if and when it is deemed necessary.

This hole, to me, is the most interesting hole we have drilled yet. The zinc sphalerite throughout the whole volcanic section occupies practically every seam or fissure available. It also has penetrated both in disseminated form and in splotchy form. This led me to a possible conclusion as to the source of the intrusion bringing the metal in. It is my thinking that the intrusion may lay below the two east-west faults, one which is in the pit and the other a half mile north of the pit. You will recall this is a highly altered fault laying between two definite faults.

Would appreciate your thoughts on this, as well as the investigation into the zinc oxide in the volcanics. If it is zinc oxide. Hole #9 becomes a minable proposition. Regards.

Sincerely,

CAC/bde

C. A. Cosgrove

P.S. I realize that there is a possibility that the green material is one of the chlorides instead of the oxide of zinc, but even if this is true there are sufficient sulphides of zinc and lead that this Hole still leaves me excited.

which I didn't want destanded prior to logging Date 11/15 /23 90402 Elevation 346 326 San Vicente ? outside of what & considered Pag (PO CH COSGROVE for his oxide. of Holes Both Holes cheserre study and 7.60 Work . Turely Hope 1954 K assays central FROM Assays - Holos 8 & 9 an ou gove Enclosed art. Copies of was taken N From Clar 84601 4 Charling 9-763 Was taken 859 East 2730 Horth Which of Took lyde Laws #9-365 2013 MESSAGE: Subject

REPLY REQUESTED --- USE REVERSE SIDE

NO REPLY NECESSARY

Rediform 45 468

Signed

SKYLINE LABS, INC. Hawley & Hawley, Assayers and Chemists Division

CERTIFICATE OF ANAI VSIS

| - | P.O. Box 50106, 1700 W, Grant Rd., Tucson, Ari | zona 85703 | ANALYSIS | | | | | | | |
|---|--|--|--------------|----------|-----------|--------------------|---------|-------|-----------|-------|
| | - SAMPLE IDENTIFICATION | GOLD | SILVER | LEAD | ZINC | COPPER | МО | * | · , | |
| 1 | CAMILE IDENTIFICATION | oz/ton | oz/ton | 1% | % | 73 | | | | |
| | • 7 · 1 Pes | Company of the Compan | reger in the | 7 | e se terr | , | star is | | | |
| | 9-365 | <0.005 | 0.19 | 0.93 | 1.36 | 0.04 | | | | |
| | # H | J. | | | | | | | | |
| | | | | | | | | | | |
| | Mr. C.A. Cosgrove | 1 9 1973 | REMA | RKS: | | CERTIFIED B | Y: | - | | |
| | 1923 W. Osborn Road Phoenix, Arizona 85015 | 1. 1.19 | Sin | gle anal | ysis | | | С | HARGES: | ¥ |
| | | Hole#9 | | 11 m | | Minimum | charge | | RATION \$ | 0.00_ |
| | MR. C.A. COSGROVE | | DATE REC | /10/73 | DATE C | OMPL.: 10/16/73 | 3 3 | 48032 | \$ 1 | 0.00 |

SKYLINE LABS, INC.

Hawley, & Hawley, Assayers and Chemists Division P.O. Box 50106, 1700 W. Grant Rd., Tucson, Arizona 85703

CERTIFICATE OF ANALYSIS

GOLD SILVER LEAD ZINC COPPER МО - SAMPLE IDENTIFICATION ZnO % % % 763 0.44 0.01 0.02 REMARKS: CERTIFIED BY:

Mr. C. A. Cosgrove

1923 W. Osborne +

Phoenix,_Arizona

cc: Santa Monica, Califorma

Single analysis

CHARGES:

PREPARATION \$

Split Core being mailed under separate cover. DATE COMPL.:

ANALYSIS \$10.

ACCT.:

10/18/73

\$10.00

MR. C.A. COSGROVE

10/12/73

348095

COPPER STATE ANALYTICAL LAB., INC.

300

DNYANENDRA A. SHAH ARIZONA REG. NO. 8888 REGISTERED ASSAYER
P. O. BOX 7517
TUCSON, ARIZONA 85725

710 E. EVANS BLVD PHONE 602-884-5811 884-5812

James Stewart Co. Attn: Harvey Hays 707 Mayer Central Bldg. 3033 N. Central Ave. Phoenix, AZ 85012

JOB # _____001454

RECEIVED ___8/27/82

REPORTED ____9/3/82

INVOICE # ____C ___1775

| Mo ppm |
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1 ppm = 0.0001%

1 troy oz./ton = 34.286 ppm

1 ppm = 0.0292 troy oz./ton

* Gold and Silver reported in troy oz. per 2,000 lb. ton.

COPPER STATE ANALYTICAL LAB., INC.

301

DNYANENDRA A. SHAH ARIZONA REG. NO. 8888 REGISTERED ASSAYER
P. O. BOX 7517
TUCSON, ARIZONA 85725

710 E. EVANS BLVD. PHONE 602-884-5811 884-5812

James Stewart Co. 707 Mayer Central Bldg. 3033 N. Central Ave. Phoenix, AZ 85012 Attn: Harvey Hays

JOB# 001493

RECE IVED 9/17/82

REPORTED 9/27/82

INVO ICE# C 1825

| /ey | Hays | #10 | HOLL |
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| AZ | 85012 | | |

| SAMPLE | li vey riays | | HULL | 1 | 1 | 1 | 1 |
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| 100 | | 62 | 20 | 160 | | | |
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