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PRELIMINARY EVALUATION OF THE  
VULTURE MINE PLACER POTENTIAL  
MARICOPA COUNTY, ARIZONA

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Date: January 23, 1985

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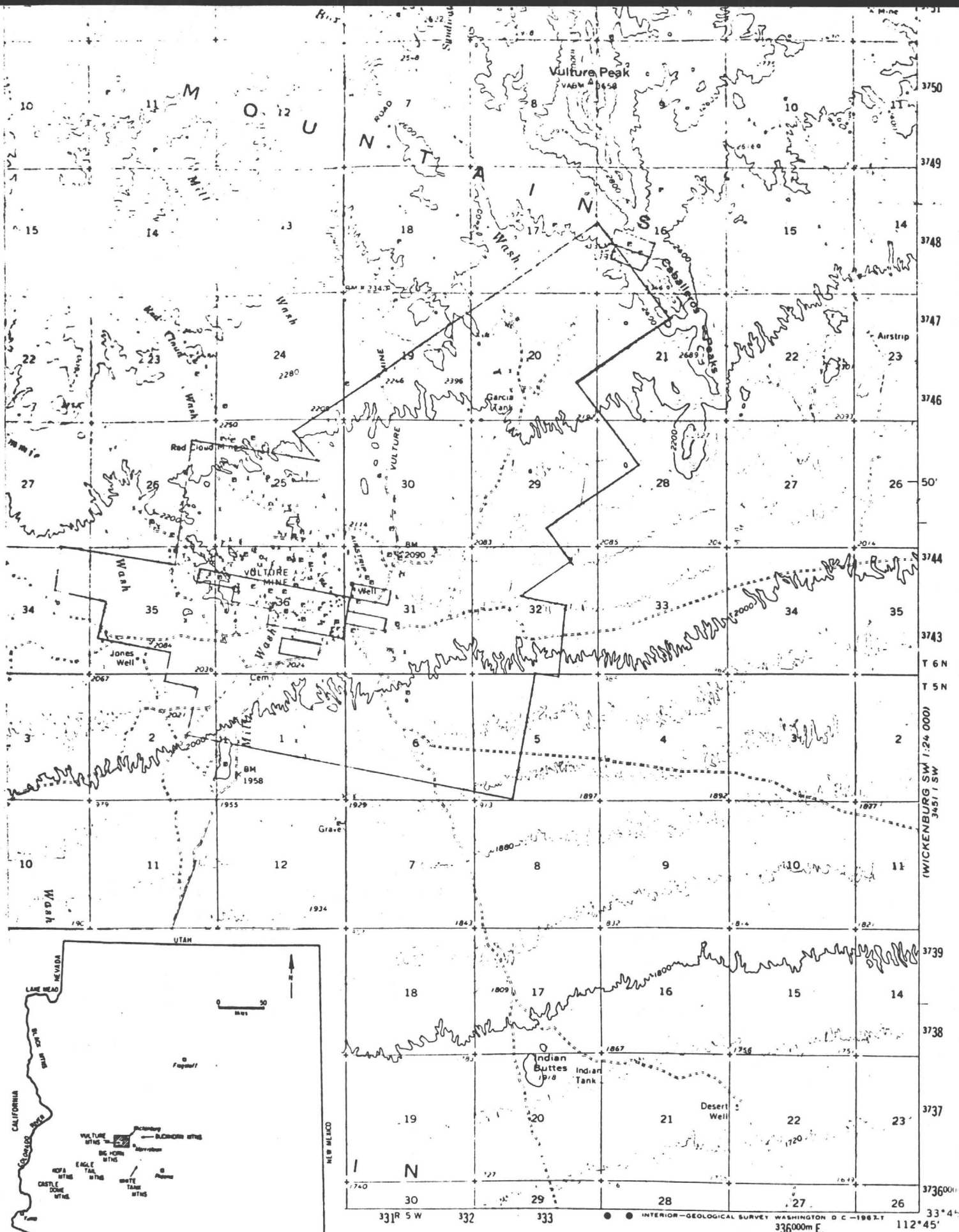
## Appendices

I Channel Sampling Geologic Mapping
II Recovered Nugget Weights
III Analytical Procedure and Assay Results
IV Concentrate geochemical Results

1. INTRODUCTION

This author was retained by DMEA, headquartered in Scottsdale, Arizona, on a fee-paid consulting retainer basis to investigate the placer potential of the Vulture Mine area. The object of this preliminary testing program was to determine the presence of placer gold in these alluvials and whether economic concentrations could exist adjacent to the Vulture Mine. This field investigation encompassed the period of December 3rd to December 20th 1984.

The Vulture Mine is located approximately 16 miles southwest from Wickenburg. The property is easily accessed by a well graded and partially paved road from Wickenburg located in the extreme northern portion of Maricopa County, Arizona. Vegetation in the area consists of cactus and assorted desert brush which flourishes at this 200 foot elevation. Stream gulleys can contain water only after severe rain storms and consequently the Vulture Mine water is obtained from deep wells at a reported pumping rate of 75 gpm.



## 2. SAMPLING

Reconnaissance evaluation of a number of potential alluvial channels is best facilitated by excavating a sequence of sample points along strategically placed "fences." Hand excavated channels within back hoe trenches provides large volume geological samples for mechanical upgrading. The following described each phase of this operation.

2-1 Each "fence" of back hoe trench sites were selected after reviewing local geology, available maps and enlarged air photograph of the Vulture Mine area. A back hoe equipped with bucket rippers and capable of an effective excavating depth of 20 feet was used to provide samples at approximately 200 foot intervals along five (5) "fences" illustrated in Figures 3 through 7. Many trenches did not reach maximum depth penetration as either bedrock or dense caliche formation effectively halted practical excavation. Photograph 1 illustrates the excavator that was used on this project.

2-2 Hand excavated channels were cut the full depth or at selected intervals in strategic trenches. The typical cross sectional dimensions were 1.5 feet x 1.0 feet, however changes in this standard procedure were dictated by specific geological conditions. Channel lengths were also adjusted to geological criterion illustrated in mapped channel profiles located in Appendix I. Statistical summary of channel sample data as follows:

- 17 trenches sampled
- 7 sampled trenches reached bedrock
- 39 channels excavated
- 6.9 cubic feet average channel volume
- 20.3 feet maximum sampled depth
- 13.4 feet average sampled depth





PHOTOGRAPH 1

Excavator used in trench digging. Note twin ripper teeth on 7/8 cu yd bucket



PHOTOGRAPH 2

Sampling plant consisting of upper Denver Gold Saver and lower E-Z Panner for fines recovery

2-3 Excavated channel samples were weighed prior to mechanical processing. Individual sample data is located in Table I, with the following averages for the project:

849 pounds/sample

123.8 pounds/cu. ft. bulk density (assuming 0% moisture)

1.67 tons/cu. yd.

The processing plant utilized consisted of a two stage process (photograph 2). A "Denver Gold Saver" emulsified, classified at 5/16 inch and provided a "rougher" concentrate. The 5/16 inch tails from this unit were classified in an E-Z Panner at 20 mesh providing the "cleaner" concentrate. The E-Z Panner typically recovered virtually all the black sand and fine gold in each sample. The E-Z Panner was used initially as the primary recovery plant in samples 1-1-1 through 1-1-3 screening at 8 mesh.

Concentrate from both units were combined and hand panned producing an 85% black sand concentrate containing all the free gold. Rejects from this process were screened at 10 mesh producing the tails samples found in Table I. Black sand concentrates averaged 805.6 grams or approximately 4.18 Pounds/ton of the feed weight.

2-4 Visual gold grain counts were performed on each black sand concentrate prior to laboratory submittal. Three samples for each processed channel were submitted to Jacobs Assay Laboratory, Tucson for analysis. The black sand concentrate was submitted for amalgamation and subsequent fire assay processing with total gold reported in milligrams. Prior to amalgamation, all nuggets (> 1.5 mm) were weighed individually with their weights located in Appendix II. Detailed amalgamation procedure is located in Appendix III. The concentrate residue was fire assayed using a two (2) assay ton sample. The tails samples were each analyzed using a

two (2) assay ton fire assay charge. Results are listed in Table 1.

Select five (5) foot interval samples from previously drilled exploration reverse circulation holes were hand panned and the resulting black sand concentrates submitted to Jacobs Assay Laboratory for total fire assaying. Analytical results, together with visual gold notations, are located in Table II.



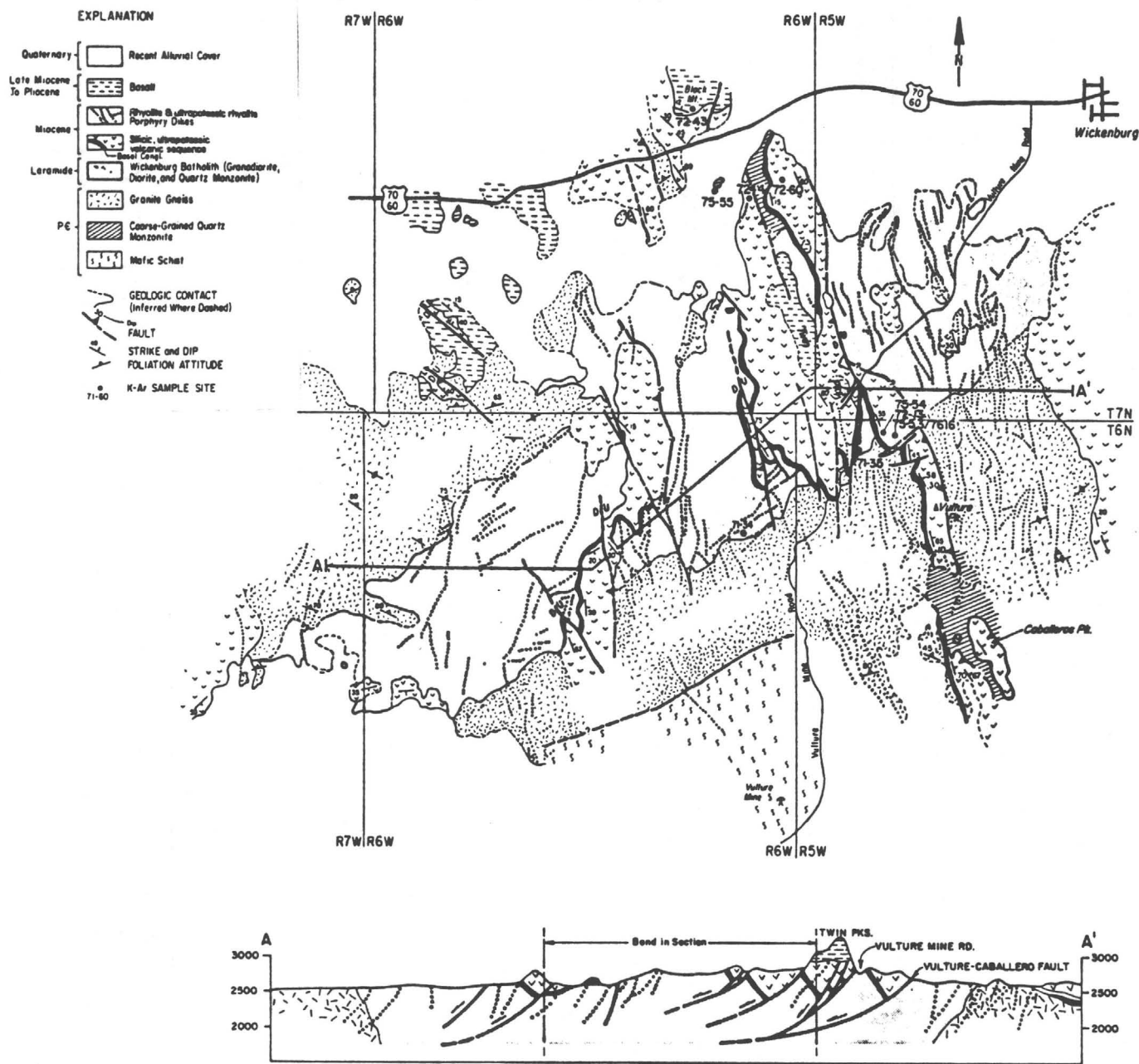


Figure 2  
Generalized geologic map and cross section of the Vulture Mts. Precambrian rocks exhibit crude northeast pattern. Erosional unconformity exists below Miocene acid volcanics and Mio-Pliocene basalts. (After Rehrig, Shafigullah & Damon; 1980)

### 3. GEOLOGY

3-1 Regional bedrock geology has been mapped by a variety of individuals. The most recent by Rehrig, Shafigullah and Damon (Arizona Geol. Soc. Digest. Col. XII, 1980) is found in Figure 2. They have summarized the geology of the Vulture Mountain area as:

"Geologic mapping and geochronologic studies in the Vulture Mountains near Wickenburg, Arizona, have lead to the recognition of a large, northeast-trending batholith of 68.4 m.y. age that intrudes complex gneissic and granitic rocks of probably Precambrian age. Overlying the denuded crystalline terrain is a sequence of late Oligocene to Miocene ( 26 to 16 m.y.) volcanic rocks (vitrophyres, ash-flow tuffs, welded tuffs, breccios,<sup>sp</sup> agglomerates and lava flows) that vary locally. ----- Overlying this volcanic sequence in angular unconformity is a thin section of basal conglomerate and basalt lava flows dated at 13.5 m.y."

Tertiary sedimentation commenced in the Eocene coinciding with the Laramide orogeny. The landscape became increasingly higher in relief producing ever increasing percentages of clastic sediments, including gravels deposited in tectonic basins. This  $\pm 30$  m.y. sedimentation phase was terminated with the deposition of the Oligocene-Miocene acid volcanism. A second phase of clastic sedimentation ensued, which was subsequently covered by basic volcanism. Miocene and Pliocene sedimentation occurred within closed basins and produced essentially evaporite and carbonate sequences. Interbedded fine grained clastic sediments were deposited

adjacent to basin margins. The development of external drainage in the Pliocene greatly diminished evaporite-carbonate deposition which evolved into well-defined fluvial systems, without ponding, in the Pleistocene. It is believed that these established drainage systems prevailed to the Recent.

3-2 The above brief and generalized chronologic Tertiary cyclic tectonics, volcanism and sedimentation had a pronounced influence on the auriferous fluvial channels exposed during this current placer evaluation program. The well-developed regional southerly trending drainage system (Figure 1) in the Vulture Mountain area probably existed for a considerable geologic period.

Detailed geologic mapping in 1984 by Noranda Exploration (unpublished map and summary report) revealed a complex sequence of east-west trending north dipping meta-sediments and volcanics dislocated by major north-northwest and north-south trending fault zones (Talmadge and Schoolhouse Faults, respectively). These intersecting structures undoubtedly greatly influence the courses of the fluvial channels in the Vulture Mine area. In fact, the position of known placers greatly assisted in locating underground structures in the Vulture Mine (E&M. J. Vol. III No. 7, 1921).

Trench mapping during this placer evaluation-sampling program indicated general bedrock slopes to the south with coincident fluvial paleocurrent directions revealing gradients of 5 to 10 percent. Mapped east-west striking north dipping ( $110^{\circ}31^{\circ}$ ) reverse faults within the fluvial sediments (Photograph 6) suggests that minor essentially bedding plane dislocations also influenced fluvial sedimentation.

Sedimentological evaluation of all fluvial sequences exposed in trenches suggest these channels occupy a relatively high profile position within the regional sedimentation pattern. Semi-angular clasts (rounding 3-5) appear locally derived and occur as moderately to well sorted and packed gravels in narrow high energy multiple sedimentation pulses. Individual channel width to depth ratios suggest a relative diminishing fluvial energy upward in these relatively unconsolidated clastic units. This correlates with the graded sequences seen in many locations. The grossly unsorted nature of the uppermost sediments suggests progressively less humid conditions later in the sedimentation pattern as reflected by the present arid climate.

The following detailed trench "fence" descriptions attempt to elaborate on local variances within the above generalized sedimentation pattern.

### 3-3 Southeastern Placers

This series of southwest-northwest trenches (Trenches 13, 1, 2 and 14) seen in Figure 3 represent mature fluvial systems typified by well-defined bedding outlining moderate energy systems deposited on irregular bedrock surfaces (Photographs 4, 5 and 6). Detailed channel sample mapping (Appendix I) well illustrates the transition from unsorted alluvium to broad low-moderate energy broad channel development to higher energy scour features near bedrock (Trenches 1, 2 and 14). Trench 13 represents a higher elevation or late phase system consisting of small sorted scour channels marginal to the main fluvial system in Trenches 1 and 2.

### 3-4 Southern Placers







PHOTOGRAPH 3

Careful attention to final  
clean-up in channels



PHOTOGRAPH 4

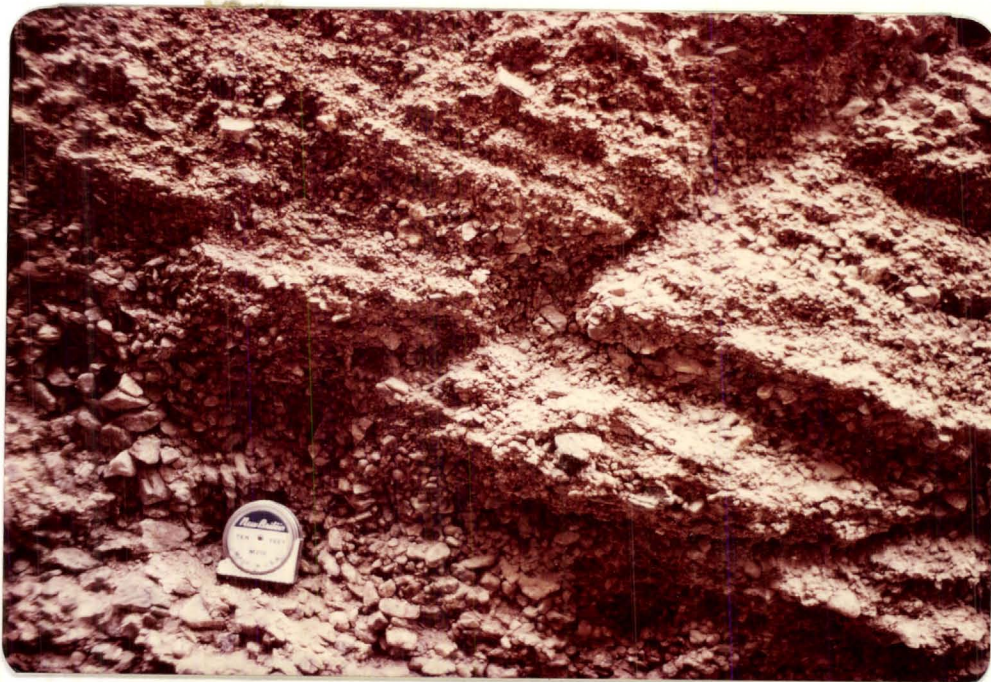
Bedrock pinnacle within  
fluvial sediments in Trench 1





PHOTOGRAPH 5

Full wall channel in 20-foot trench. Yellow ribbons mark channel limits



PHOTOGRAPH 6

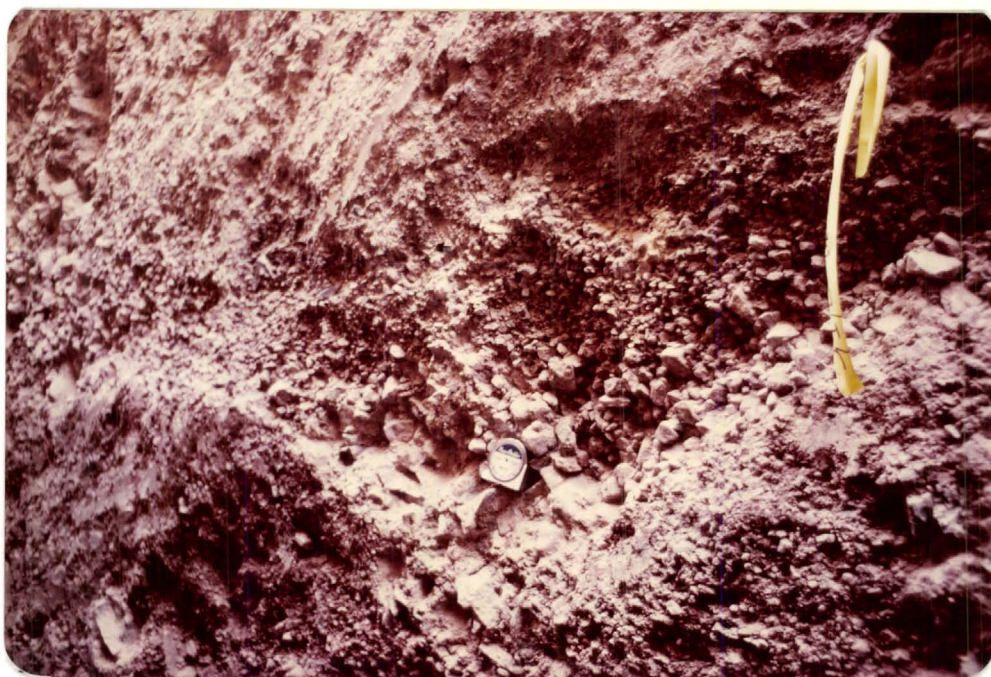
Low angle small scale reverse faulting displacing bedded and sorted fluvial sediments in Trench 1





PHOTOGRAPH 7

Trench 4 illustrating yellow mine tailings overlying scour bedding with increasing conglomerate percentage downward



PHOTOGRAPH 8

High energy well sorted channel adjacent to sample 3-1-2



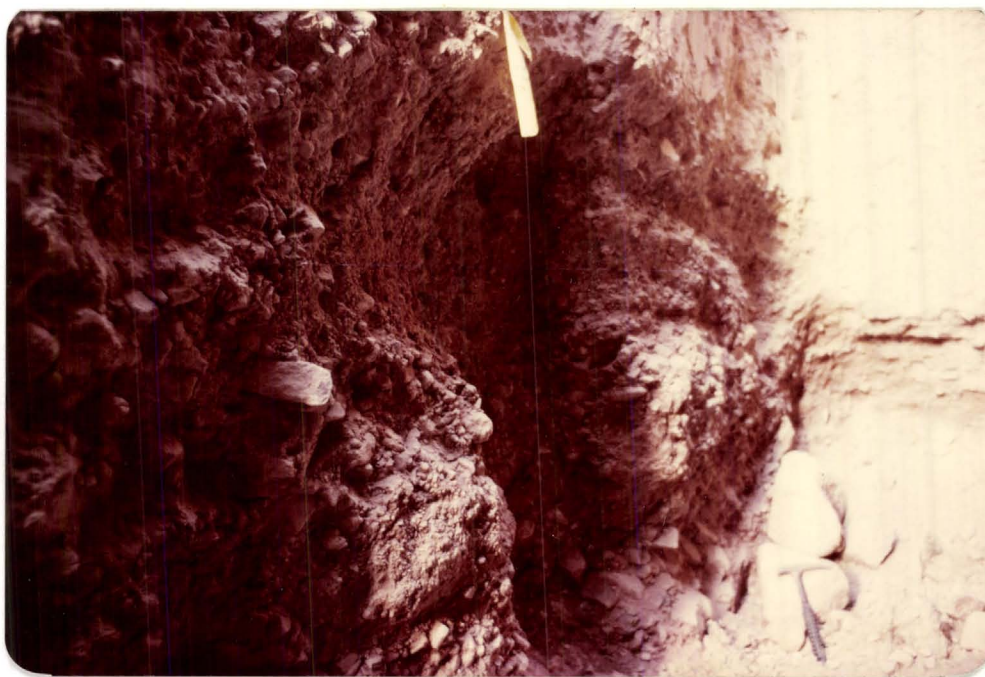


Figure 4  
Vulture Mine Tailing Area showing tailings isopachs based on  
drill holes and proposed pre-tailings erosion surface channel axes



PHOTOGRAPH 9

Sample 3-1-2 illustrating moderately sorted and  
and graded fluvial conglomerate overlying bedrock



PHOTOGRAPH 10

Base of Trench 7 illustrating semi-angular pebbles within  
a small pebble-grit unconsolidated fluvial sequence





A series of east-west trenches underlying existing mine tailings reveal sequences of high energy fluvial activity partially exposed in Trenches 3 and 4 (Figure 3 and Appendix I). Graded bedding is very evident in this area with moderately rounded cobbles located close to bedrock as generally poorly sorted gravels (photographs 4 and 5). Well sorted and packed scour channels occur at various levels within the basal 50% of this sequence (photograph 7). Bedrock exposed in Trench 3 (Photograph 8) would indicate irregular scouring activity probably controlled by both bedrock lithology and structure.

Contouring overlying mine tailings from drill records (Figure 4) suggests a north-south and northwest-southeast erosion surface channel network centered on the Vulture Mine outcrop system. Angular quartz pebbles in the placer gravels indicate a very local source. Trench bottom bedrock exposures reveal a westward 13:1 bedrock slope from Trench 3 to Trench 4, suggesting the main drainage axis possibly correlates with a larger paleo drainage system as indicated in Figure 4.

### 3-5 Eastern Placer

A series of east-west trenches in the Air Strip area disclose relatively deep fluvial channels which must be deeply incised into bedrock as Trenches 5, 6, and 7 (Figure 5) failed to reach bedrock, although schistose bedrock is exposed in shallow flanking trenches. Graded bedding is definitely seen in Trenches 5 and 7 where multiple well-sorted and bedded clastic sediments become progressively coarser grained at depth. The 15:1 width to depth ratio for Trench 5 reveals a moderate fluvial energy system corresponding to a well-developed rhythmic bedded

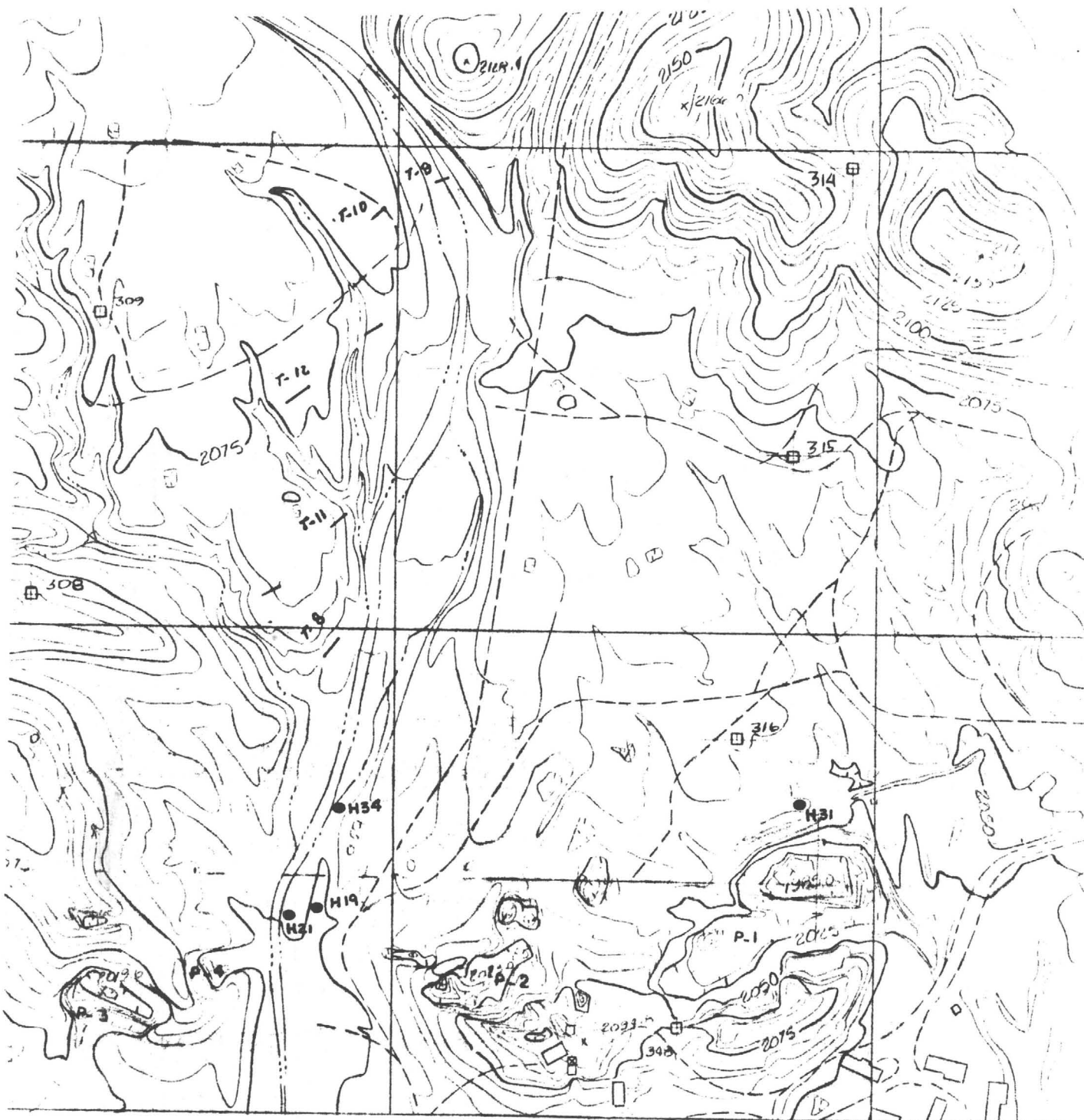


Figure 6  
Vulture Mine Placer Project location of northern area sampled  
trenches and drill holes, T-9 sampled trench, H34 sampled drill hole





PHOTOGRAPH 11

Sample 11-1-4 illustrating the upper well sorted fluvial channel overlying well bedded poorer sorted small pebble conglomerate



PHOTOGRAPH 12

Samples 12-1-2 and 12-1-3 illustrating graded moderate energy fluviate channel development

sedimentation pattern containing well-packed robust gravels at the bottom. A well-sorted fine grained sand overlies well-bedded and graded coarse fluvial gravels in Trench 7 (Appendix I and Photograph 10). The exposed alluvial section in Trench 6 suggests these low energy sediments occupy a relatively high level fluvial sequence relative to the more basal sections in Trenches 5 and 7.

### 3-6 Northern Placer

A total of five (5) trenches (Figure 6) were sampled in an area of broad but dissected alluvium north of the Vulture Mine. Four (4) of these excavations reached bedrock (Appendix I) and one (Trench 9 - not mapped) was cut in a gully containing active stream wash. All mapped trenches revealed progressively increasing fluvial sedimentation with depth reflected as graded bedding, increased sorting and packing. Relatively dense caliche was encountered cementing near surface rubble.

Mapping Trench 8 revealed a bedrock relief of at least two (2) feet overlain by a well-packed poorly sorted conglomerate containing metamorphic, igneous and conglomerate clasts. Trench 9 revealed a well-packed graded sequence with sub-rounded pebbles to 6 cms lying on a schistose bedrock. A similar coarse robust clastic sequence was evident on bedrock in Trench 12 (photograph 10). However, multiple scour channels containing small pebbles and grits were evident in the middle sections suggesting a change from relatively high energy robust fluvial action to overlying lower energy pulsating sedimentation. Although Trench 11 bottomed on hard caliche (false bedrock?), graded bedding with well-packed and sorted gravel lenses are very prevalent and suggest a modest energy

fluvial system (photograph 11).

### 3-7 Southwestern Placer

Trenching in this area was severely restricted by a thick layer of near surface hard caliche cemented cobble wash. Two trenches reached maximum practical excavation depth (Figure 7 and Appendix I) and will be discussed at length.

Trench 15 intersected multiple gravel horizons depicting pulsating fluvial sedimentation depositing rather thin pebble horizons followed by low energy weakly bedded pebbly sands. The igneous (dioritic) clasts are better rounded than the metamorphics probably reflecting durability factors rather than variable transport distance. Clast rounding is markedly better in the sampled portion than in the near-surface unsorted gravels.

The fluvial section exposed in Trench 16 contrasts greatly with Trench 15, as the clasts are nearly 100% porphyritic quartz diorite and more angular. A scattering of the greenish metamorphic clasts were seen in the excavation. The very uneven bedrock is essentially this same igneous rock, suggesting a very local source for this sediment. A relatively steep paleocurrent gradient (approximately 11%) coincides with the multiple narrow cut and fill high energy channel development.



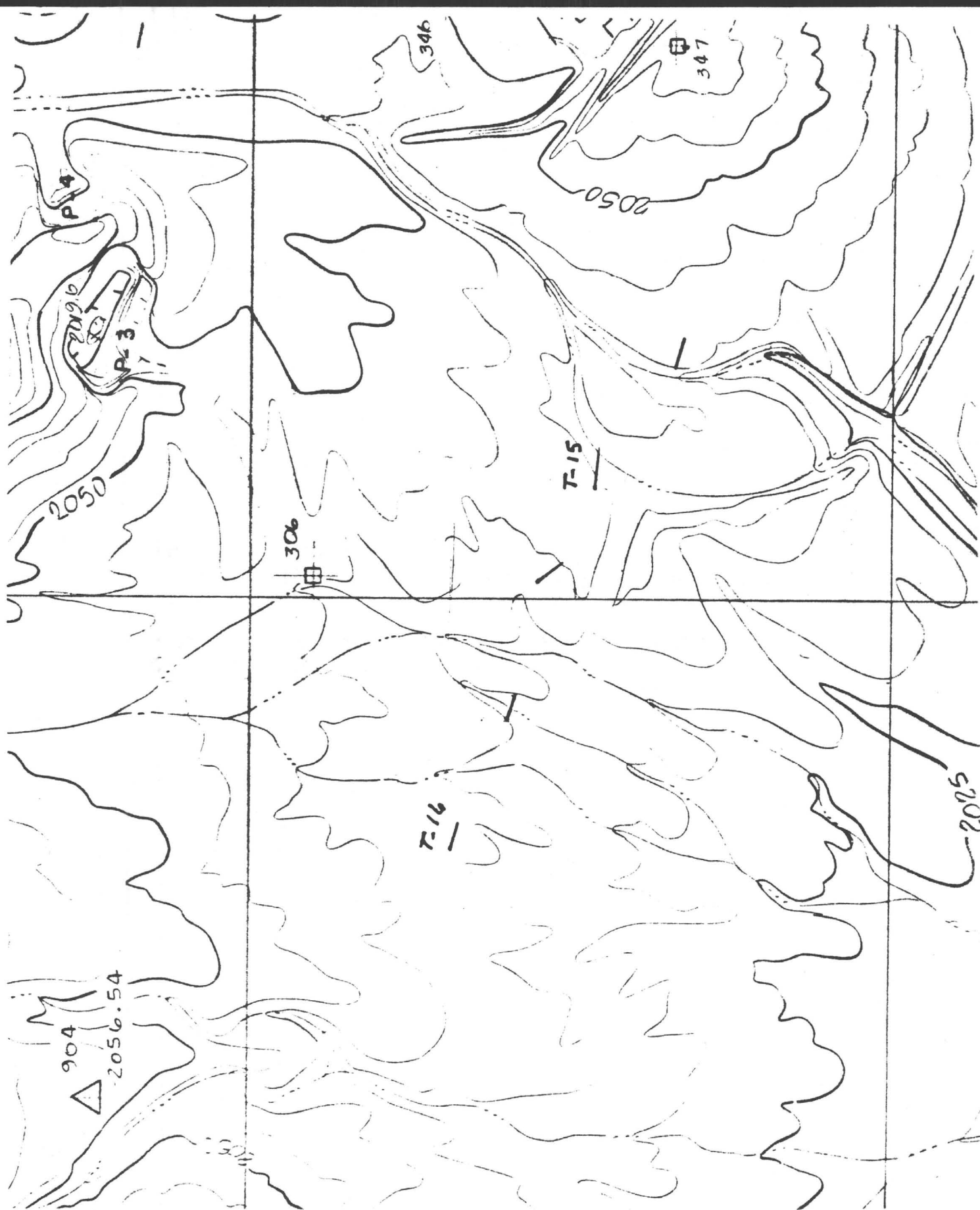


Figure 7  
Vulture Mine Placer Project  
Showing sampled trenches in southwest area.

VULTURE MINE PLACER SAMPLE DATA TABLE I

Sample No.	Depth Below Surface	Sample Wt. (lb.)	Sample Vol. (ft <sup>3</sup> )	Con. Wt. (g)	T. F. lb/cu.ft	Gold Nugget	Particle #1	Size Distribution #2	Distribution #3	Fines	Amalgamation Au mg.	Tails OPT	Plant Tails +10# Au	Plant Tails -10# Au	Placer Grade mg/cu.yd
1-1-1	1.5'-7.5'	909	12.00	927	75.75						6.734	.005	.009	.001	15.15
1-1-2	7.5'-11.5'	730	8.00	673	91.25						2.780	.007	.002	.002	9.38
1-1-3	11.5'-16.0'	1431	9.00	4635	159.00				1		13.145	.002	.001	.001	39.44
1-2-1	3.0'-7.5'	805	9.00	503	89.44						0.835	.010	.003	.002	2.51
1-2-2	7.5'-12.0'	718	8.00	399	89.75						3.210	.014	.002	.001	10.83
2-1-1	2.0'-6.5'	973	6.75	722	89.75						0.765	.002	.003	.002	3.06
2-1-2	6.5'-12.5'	1087	9.00	1126	144.15			2			9.220	.004	.001	.002	27.66
3-1-1	6.0'-11.5'	1082	8.75	1643	123.66						1.990	.013	Tr	Tr	6.14
3-1-2	11.5'-16.5'	1069	6.50	553	164.46	2	1	15			451.100	.023	.002	.002	1874.22
3-2-1	11.3'-16.5'	955	7.75	612	123.23	1	3		2		107.310	.033	.005	.001	373.85
4-1-1	2.0'-7.2'	753	7.75	424	97.16						3.410	.045	Tr	Tr	11.88
4-1-2	7.2'-11.3'	747	6.25	413	119.52		1	16	10		20.180	.013	.002	.001	87.18
4-2-1	7.3'-11.3'	757	6.00	468	126.17			5	7		11.950	.050	Tr	.002R	53.78
5-1-1	1.0'-4.0'	367	4.50	271	81.56						0.040	.010	Tr	.001	0.24
5-1-2	4.0'-11.0'	1417	10.50	780	134.95						0.153	.037	Tr	Tr	0.65
5-1-3	11.0'-17.0'	1268	9.00	931	140.89						0.062	.003	.001	Tr	0.19
6-1-1	0-6.0'	840	9.00	1337	93.33				1		0.140	.010	.001	.001	0.42
6-1-2	6.0'-11.0'	871	7.50	637	116.13						0.580	.011	Tr	Tr	2.09
7-1-1	1.0'-6.0'	283	3.75	70	75.47						0.030	.528/.583R	.002	Tr	0.22
7-1-2	6.0'-10.25'	273	3.19	86	85.58						0.063	.096/.091R	.005	.002	0.53
7-1-3	10.25'-15.0'	848	7.00	949	121.14			2			0.022	.005	.002	.002	0.08
8-1-1	0-4.0'	667	4.00	349	166.75						0.026	.006	.002	.002	0.10
8-1-2	4.0'-8.0'	777	4.00	755	194.25						0.468	.006	.003	.002	3.16
8-1-3	8.0'-13.0'	701	4.83	461	145.13				1		14.530	.020	.001	.004	81.22
9-1-1	0-4.5'	771	6.50	1124	118.62		1	1			18.020	.031	.001	.001	74.85
10-1-1	0-4.33'	971	6.50	824	141.08				1		11.054	.014	.002	.008R	45.92
11-1-1	3.0'-6.7'	413	4.13	199	100.00						0.413	.008	.001	.002	2.70
11-1-2	6.7'-12.0'	973	8.00	784	121.63						3.180	.009	.002	.001	10.73
11-1-3	12.0'-16.3'	874	6.50	867	134.46			1	1		0.835	.002	Tr	.001	3.47
11-1-4	16.3'-20.3'	756	6.00	385	126.00			1	3		5.553	.014	.001	.004	24.99
12-1-1	4.0'-8.0'	796	6.00	587	132.67				1		0.434	.006	Tr	Tr	1.95
12-1-2	8.0'-12.0'	828	6.00	569	138.00						1.430	.002	.001	.002	6.44
12-1-3	12.0'-15.5'	799	5.25	591	152.19				2		0.838	.004	.001	.001	4.31
13-1-1	0-4.9'	568	4.79	289	118.58	1	2	4	10		66.650	.663/.467R	.002	.003	375.69
14-1-1	9.0'-15.0'	1026	9.00	1195	114.00						2.455	.003	.004	.001	7.37
15-1-1	8.0'-11.7'	620	5.50	2480	112.73						0.179	.005	.001	Tr	0.88
15-1-2	11.7'-16.0'	990	6.50	1350	152.31						14.300	.002	Tr	.001	59.40
16-1-1	7.0'-11.0'	623	6.00	247	103.83				1		1.800	.014	.001	.001	8.10
16-1-2	11.0'-17.0'	1029	9.00	204	114.33		2	1	20		74.160	1.028/1.133R	.002	Tr	222.48

#### 4. RESULTS

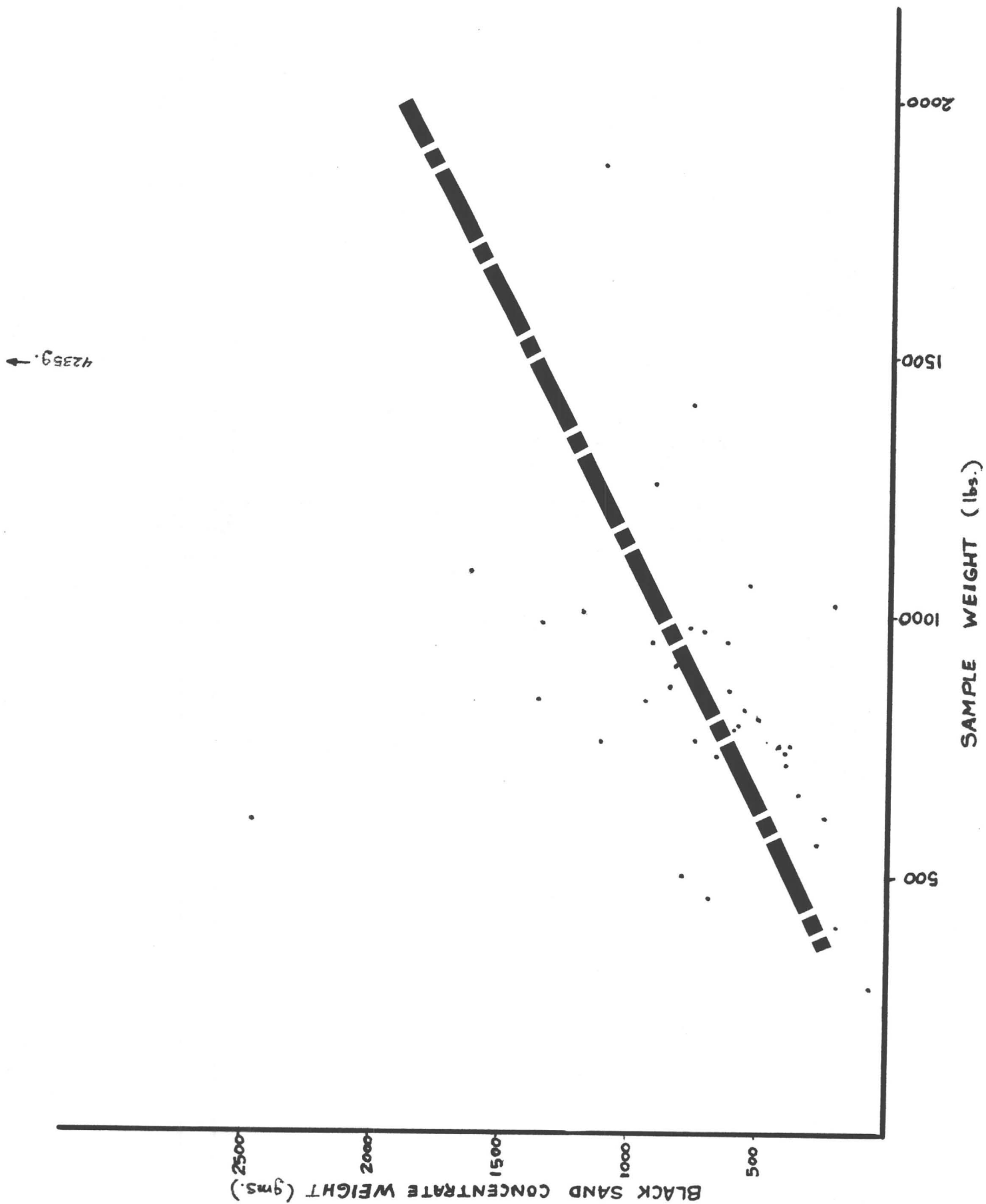
Evaluation of placer deposits is frequently haunted by the well-known particle sparsity effect. Sample processing efficiency can be yet another critical concern in determining the economic viability of a property. Sedimentological aspects of placer geology frequently explains the seemingly random and high gold distribution found in many placers. The following discusses each aspect in some detail striving to rationalize area evaluation.

4-1 Sample processing consistency can be considered good when viewing the linear relationship between concentrate weight vrs. sample weight in Figure 8 indicating a consistent relationship in recovery efficiency. Two samples (1-1-3 and 15-1-1) deviate from this well-defined trend, both of which contained abnormally high concentrate weights but low gold values.

4-2 Bulk density calculations (Table I and Appendix I) indicate that the gravel becomes more dense with depth reflecting compaction/packing density coinciding with the general increased fluviate sedimentation energy at depth. Minor amounts of caliche apparently only subtly effect bulk density levels, with the exception of Trench 12. The high bulk densities in Trench 8 are attributed to the well lithified nature of these sediments. It is interesting to note the sample that contained very high concentrate weights also correlates with a high bulk density (e.g. 1-1-3:4635 gms; 159.00 lb/cu. ft.) respectively.

4-3 Assay results are found in Table II and Appendix III along with general amalgamation-assaying procedures. Silver is not reported for the amalgamation step due to the pre-amalgamation (nitric acid) treatment of

FIGURE 8



the concentrates. Silver results for all fire assays are found in Appendix III, however are not reported in Table I due to their relatively low values. Each "fence" area will be discussed individually similar to Chapter 3 striving to evaluate economic significance to the large volume essentially geochemical samples.

#### 4-3-1 Southeastern Placers

A total of four (4) channels were cut in three (3) trenches accounting for eight (8) large volume samples in this ±240-foot-wide channel. Trench 14 is considered to occur in a parallel but separate fluvial system. Geologic mapping (Appendix I) has indicated general increased fluvial energy with depth correlating with a commensurate increase in gold content. Samples 1-2-2 and 13-1-1 consist of multiple high energy fluvial channels deposited on bedrock marginal to the main channel. The disparity in gold content (1-2-2, 10.83 mg/cuyd vrs. 13-1-1, 375.69 mg/cuyd) relates partially to the probable reworking of gold in the younger (13-1-1) channel relative to the main channel. A very interesting and significant feature is the grade increase factors for 1-2 and 2-1 which should be located over this main channel axes. Should this significant multiplier continue (4.3:1 for 1-2 and 9.04:1 for 2-1). Then this channel could contain economic gold values for a significant portion of its several 1,000 foot strike length.

The one sample in Trench 14 (14-1-1) signifies that this auriferous low energy fluvial sedimentation constitutes background level values for upper profile sediments.

#### 4-3-2 Southern Placers

Underlying the existing Vulture Mine tailings is a probable network of juvenile fluvial channels at least partially derived from the Vulture Mine outcrop system. Numerous references refer to the very bold pre-mining outcrop character of the Vulture Mine vein. Calculations indicate that the 45° dipping vein would liberate 13,575 ounces per vertical foot of erosion (e.g. 1,000' long x 32' wide x sec 45° x 0.30 opt Au) a large part of which would have been transported southward in the paleo drainage system.

Two trenches penetrated the mine tailings revealing a moderate-high energy fluvial system containing very encouraging gold concentrations which correlate with sedimentation characteristics and provenance area. Values are tabulated for Trenches 3 and 4 (Table I and Appendix I) and will be discussed in detail below:

	Total	Nugget	Residual	% Total
3-1-2	1843.8 mg/cuyd	859.86 mg/cu yd	983.94 mg/cuyd	53.4
3-2-1	<u>373.85 mg/cuyd</u>	<u>314.06 mg/cuyd</u>	<u>59.79 mg/cuyd</u>	16.0
Ave.	<u>1123.83 mg/cuyd</u>	<u>586.96mg/cuyd</u>	<u>521.87 mg/cuyd</u>	46.4

It becomes apparent from the above calculations (3-2-1 is a duplicate sample from opposite side of the trench from 3-1-2) that the 521.87 mg/cuyd average without nuggets is an attractive and probable economic placer grade with an approximate 1:1 strip ratio (tails negated). Nuggets will be a significant and real contribution to grade.

Trench 4 is approximately 170 feet west and within the same fluvial system as Trench 3 (Figure 3 and 4). Sample values from adjacent duplicate

samples are compared below:

4-1-2	87.18 mg/cuyd	
4-2-1	53.78 mg/cuyd	No nuggets recovered
Ave.	70.48 mg/cuyd	

Careful inspection of geologic mapping in Appendix I will reveal that these samples did not reach bedrock. Also, a concentration ratio of 1:7.3 between 4-1-1 and 4-1-2 indicates very significant accumulation of gold with depth suggesting continuing concentration ratios downwards with 3-1-2 and 3-2-1 level values likely on bedrock.

A random bulk sample from Trench 3 spoils was processed in the existing plant (Photograph 2). The magnetic fraction was removed and the resultant concentrate submitted for a geochemical scan. This analysis (Appendix IV) confirmed the presence of unoxidized galena seen in Trench 3 and 4 concentrates. A whitish-blue fluorescent mineral, tentatively identified as scheelite, is probably barite and zircon based on these analysis. The significance of this analysis again suggests the Vulture Mine outcrop is the primary provenance area for this placer as galena was one of the prime ore minerals obtained from the Vulture Mine. This concentrate also contained significant quantities of metallic gold ranging in size from 3 mm to -150 mesh.

#### 4-3-3 Eastern Placers

Geological mapping in this area (refer to Chapter 3-5) indicates that well-incised fluvial systems of moderate energy level occur in the three sampled trenches. Channel mapping (Appendix I) reveal graded bedding culminating in well-developed gravels at the base of Trenches 5 and 7.

Values reported for these samples are the lowest for all placers

sampled in the Vulture Mine area and a concentration ratio increase with depth is not evident. The gold grains recovered are very small and dispersed.

#### 4-3-4 Northern Placer

These five (5) trenches and four (4) drill holes (Table II) effectively sampled the fluvial profile immediately north of the Vulture Mine. Significant gold increases with depth were registered for Trench 8 and to a less extent in Trench 11. Particulate gold seen in these samples (refer Gold Particle Size Distribution, Table I) reveals that the particulate gold is almost universally fine grained and very juvenile with angular flat and wire forms predominating. Some of these gold grains were sufficiently thin to overcome surface tension/specific gravity relationships and floated during the concentrate clean-up operation.

Gold recovered in Trench 9, which was excavated to bedrock in active stream wash, revealed two larger gold grains that were moderately well rounded and coated with iron and calcium. The accompanying small grains were similar to the gold recovered in the adjacent Trench 10 which was excavated in well-packed older gravels lying directly on bedrock. The conclusion from the above reveals that the coarser gold in Trench 9 was derived from a more distant source than the small angular grains which are more characteristic of this immediate area.

Select drill cuttings from previously drilled reverse circulation holes were hand panned and the black sand concentrate fire assayed. Scattered very small angular gold grains were visible in some samples (Table II). However, value levels were very low and further indicates the



TABLE II

VULTURE MINE PLACER SAMPLES  
DRILL CUTTINGS

Hole No.	Location	Conc(g)	Au (mg)Ag		Comments
H19 0-5	26,418N/20,846E	7.0	.012	.30	1 Au grain
5-10		2.0	Tr	.20	
10-15		45.0	.110	.20	
15-20		18.0	.02	.25	
H21 0-5	26,418N/20,783E	NA	.017	NA	
5-10		NA	.058	NA	
H31 5-10	16,666N/21,840E	NA	.081	NA	1 Au grain
10-15		5.0	.124	.35	
15-20		1.5	.050	1.85	
H34 10-15	26,633N/20,880E	NA	.026	NA	2 Au grains
15-20		40.0	.005	.35	

local provenance area for these sediments contained only fine-grained free gold. The seemingly high silver values in the concentrates could benefit "hard rock" geochemical prospecting.

#### 4-3-5 Southwestern Placer

The geological contrast between the two deep trenches in this area (Chapter 3-7) is further enhanced by their value distribution noted in Table I. It becomes readily apparent that the angular and platy juvenile character of the recovered gold is entirely compatible with the host semi-angular dioritic sediments. This suggests a very local source -- either the Vulture Mine workings or some other provenance location -- for these values.

Value concentration ratios with depth are impressive for both trenches and indicates a fairly active sedimentation history for these southerly trending paleo channels. The steep paleocurrent gradient and multiple robust gravel formation (Appendix I) also attests to a vigorous sedimentation environment supplied from an adjacent provenance zone. The amalgamation tail assays would indicate that very fine grained gold is still contained within some heavy mineral.

## 5. CONCLUSIONS

The wide range of fluvial sedimentation seen during this reconnaissance evaluation of the Vulture Mine property generally consists of well incised linear channels developed relatively high on the fluvial paleoslope. These drainage patterns were probably formed at the close of the Pliocene and reached their present state of maturity during the Pleistocene humid climatic cycle. These alluvial systems degenerated with increasing aridity to the present day narrow incised gullies that are not appreciably reworking the eroding older gravels. Gold distribution profiles adequately reflect this sedimentological pattern with the very high value concentration ratios and coarser gold distribution favoring bedrock locations. Placer gold particle distribution and related values obtained from these 39 channel samples strongly suggest the Vulture Mine outcrop system was the dominant provenance area for the resulting alluvials. The measured southerly paleocurrent direction suggests the important placer concentrations will be located south of the Vulture Mine -- which is, in fact, the case. Significant quantities of fine grained angular gold in the Northern Placer area (Chs. 3-6 and 4-3-4) indicates that a secondary provenance is very likely north of the Vulture Mine.

This geological-sedimentological evaluation combined with careful sampling has revealed a significant placer potential for the Vulture Mine measuring in the 100,000's cubic yard range within the trenched areas. Specific areas containing potential economic gold values (say +200 mg/cu yd) or important geochemical concentrations will be defined based on geological deduction, value distribution and inferred economic considerations.

## 6. RECOMMENDATIONS

Specific areas of interest lie south and adjacent to the Vulture Mine outcrop system. Each area will be discussed briefly in order of importance striving to formulate objective economic decision making goals based on reconnaissance sampling.

### 6-1 Southern Placers

Channel samples in two (2) trenches revealed very attractive economic grades on bedrock (3-1-2 and 3-2-1 avg 112.3:83 mg (cu yd) and highly anomolous values (4-1-2 and 4-2-1; ave 70.48 mg/cu yd) in mid level gravels. Figures 3 and 4 indicates a pre-tailings erosion basin containing an inferred area of 15.2 acres supplied by Vulture Mine outcrop material as indicated by geologic observations and abundant galena in plant concentrates (Appendix IV).

Approximately 46.4% of the average value from Trench 3 bedrock channels (e.g. 521.87 mg/cu yd) are contained within the -1.5 mm gold particle size. These channels were approximately five (5) feet long, indicating a 1:1 strip ratio discounting overlying potentially economic mine tails.

The inferred volume of 120,000 cubic yards combined with a conservative grade of 500 mg/cubic yard based on these preliminary sampling results, indicates this area could provide economic grade gravel to a modest size placer plant. Reconnaissance drilling would be warranted to substantiate this preliminary geologic conclusion and could expand reserves

in a southerly direction.

#### 6-2 Southwest Placer

The relatively isolated position of Trench 16 and its differing sedimentological-geological characteristics compared to other fluvial systems precludes definite grade-volume estimations. However, the approximate 2:1 overburden to "ore" strip ratio is a positive feature along with the abundance of fine gold particles facilitating probable higher confidence level grade determinations (e.g. subdued particle sparsity effect). In essence, this area must be considered a second priority but important placer drilling target.

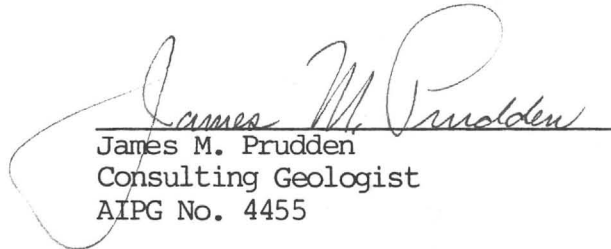
#### 6-3 Southeastern Placer

This third priority selection is based on its southerly drainage position relative to the Vulture Mine and the potentially economic grade channel in Trench 13 indicating that economic bedrock concentrations could exist in this area. The numerous intersecting trenches suggest a channel width  $\pm 240$  feet and approximate maximum depth of 30 feet (never reached by the back hoe). Considering the basal five (5) feet could be economic (5:1 strip ratio) then this channel could produce say 45,000 cubic yards per linear 1,000 feet. Select drilling is required to confirm the above geologic prediction.

The above three (3) prioritized placer drilling targets could, in their entirety, produce an attractive modest (say 150 cuyd/hr.) placer operation. Obviously, the southern placer area has the highest probability of producing an economic placer on its own merits. The overlying possibly economic mine tailings must also be considered in the economic feasibility

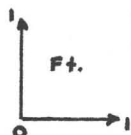
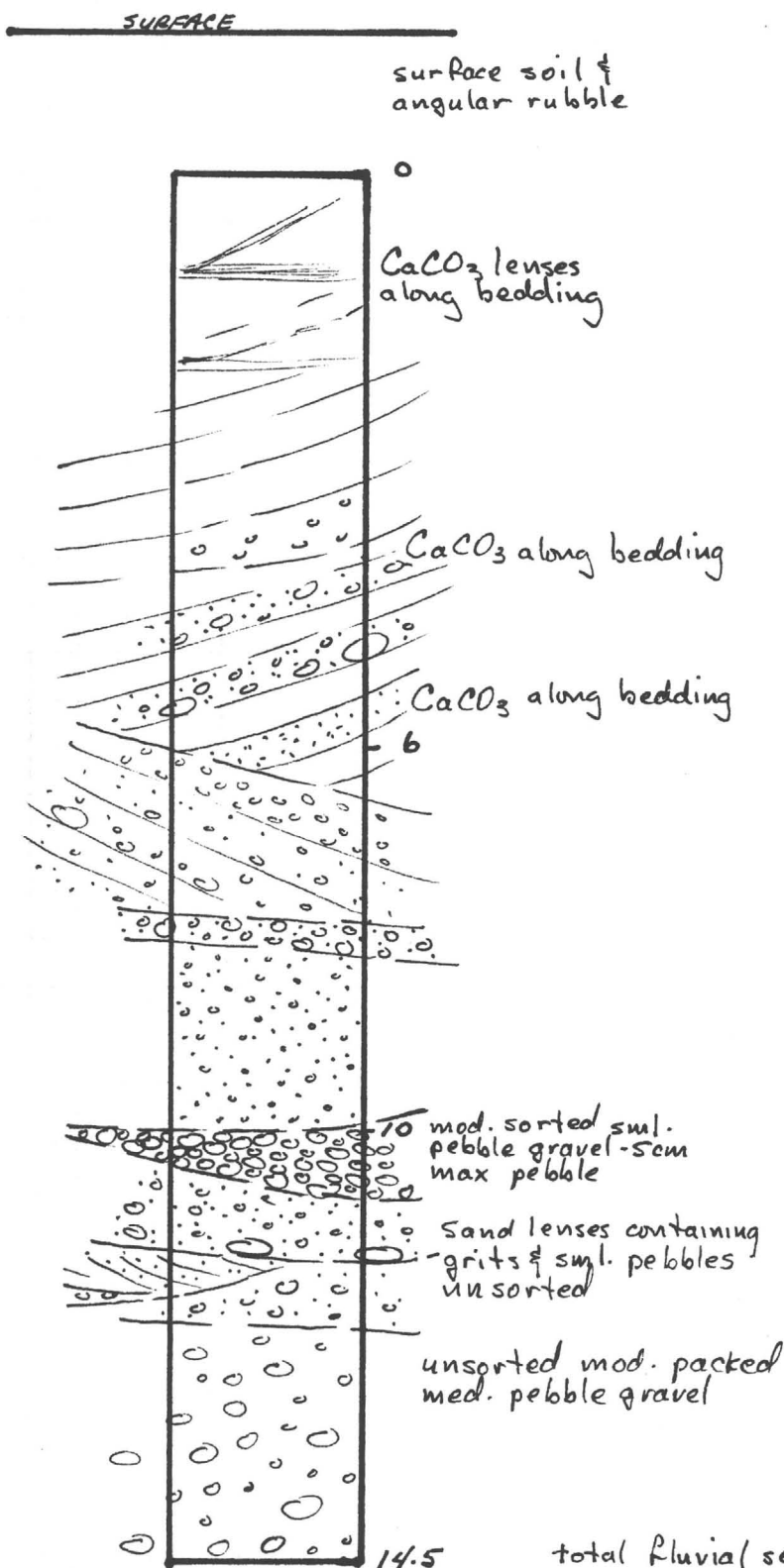


study for this small area. A modest drilling program could very well expand the economic scope for this historic property to the position of a modest placer gold producer for a substantial period.

  
James M. Prudden  
Consulting Geologist  
AIPG No. 4455

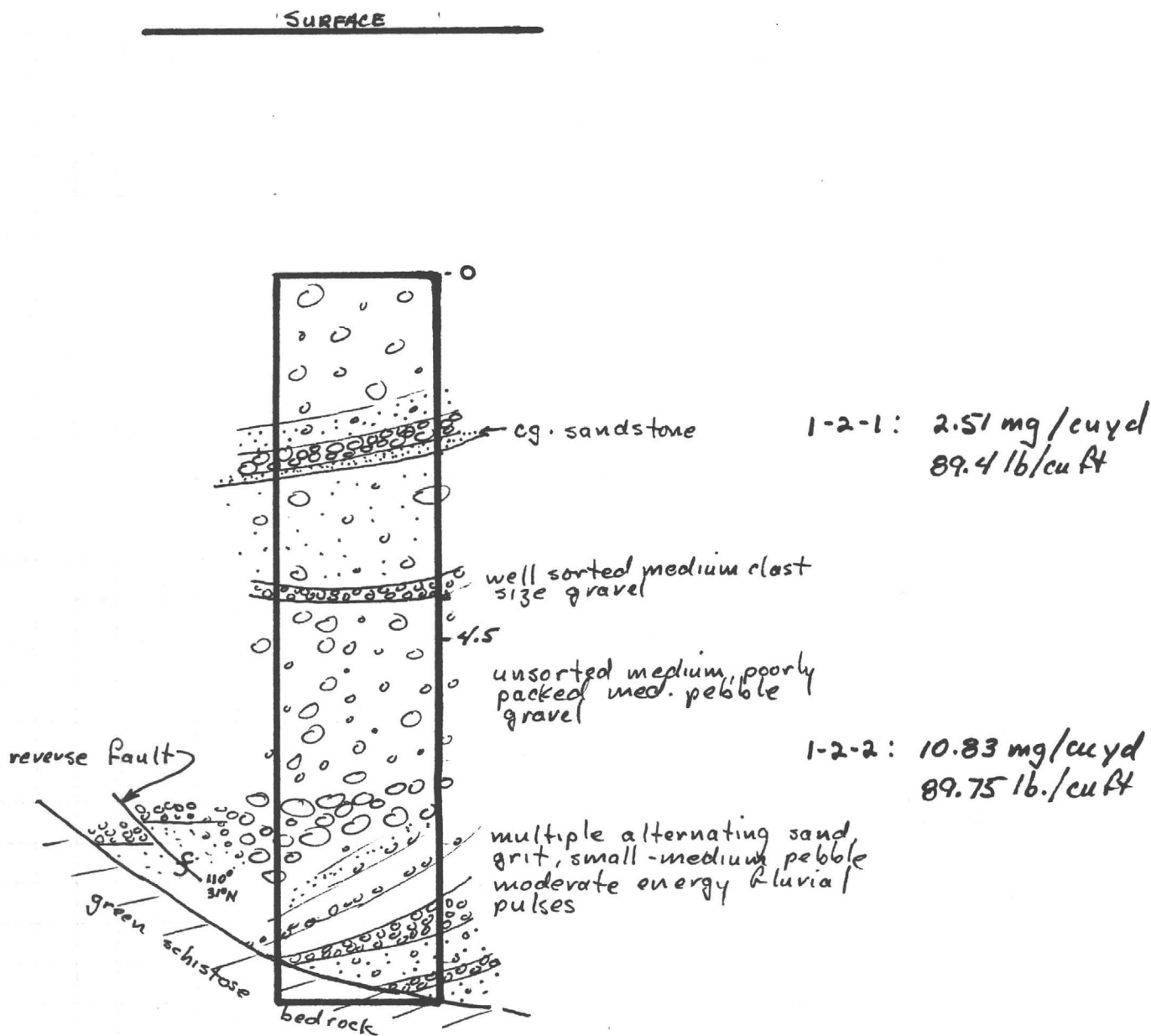
APPENDIX I

VULTURE MINE  
PLACER PROJECT  
TRENCH 1  
CHANNEL 1



total fluvial sequence graded, most individual sed. phases mod-well sorted; clasts semi-angular, rounding 3-4; 70% clasts green schistose metamorphics to 35cm, 28% clasts igneous & 2% clasts basalt

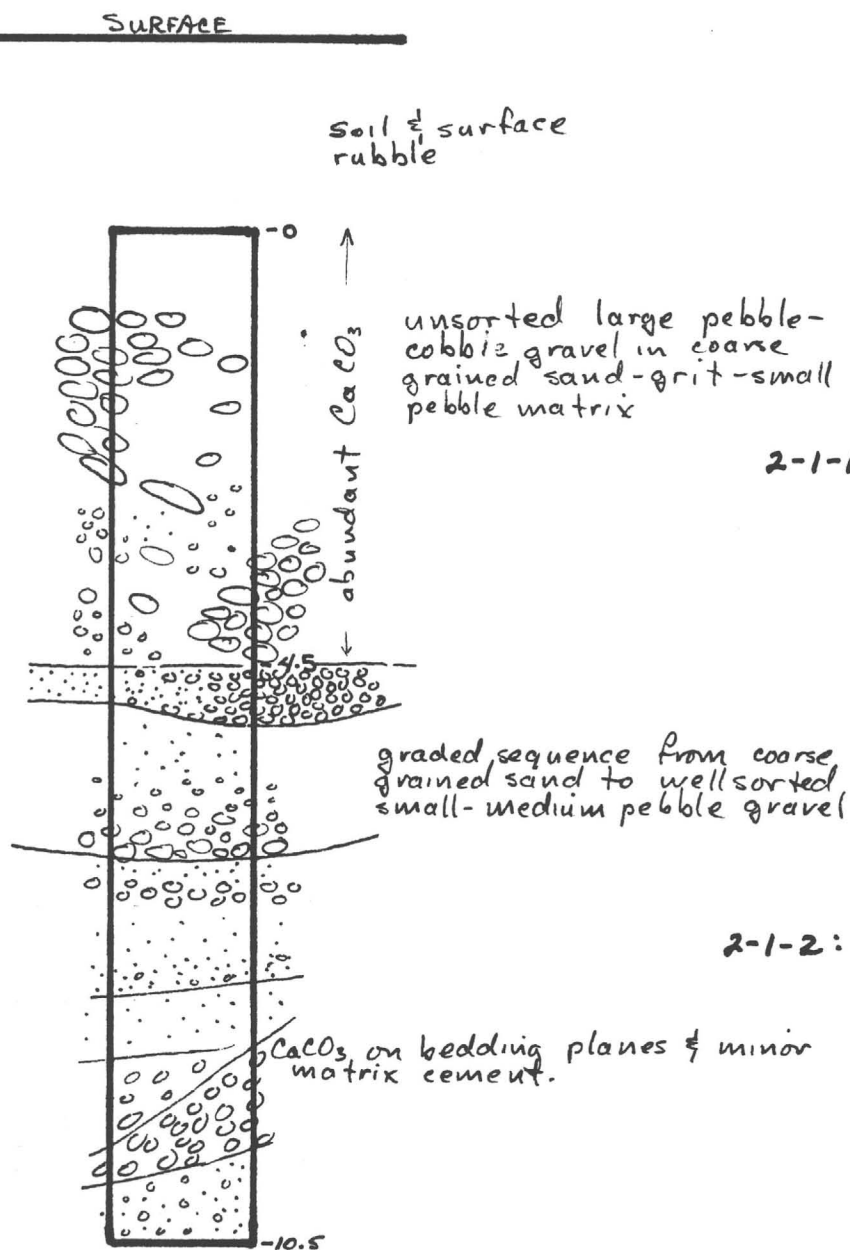
# VULTURE MINE PLACER PROJECT TRENCH 1 CHANNEL 2



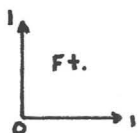
similar fluvial sequence to channel 1.  
bedrock influence on sedimentation very evident, clastic matrix coarse grained, semi-consolidated containing only minor clay; scour troughs 185° azm.



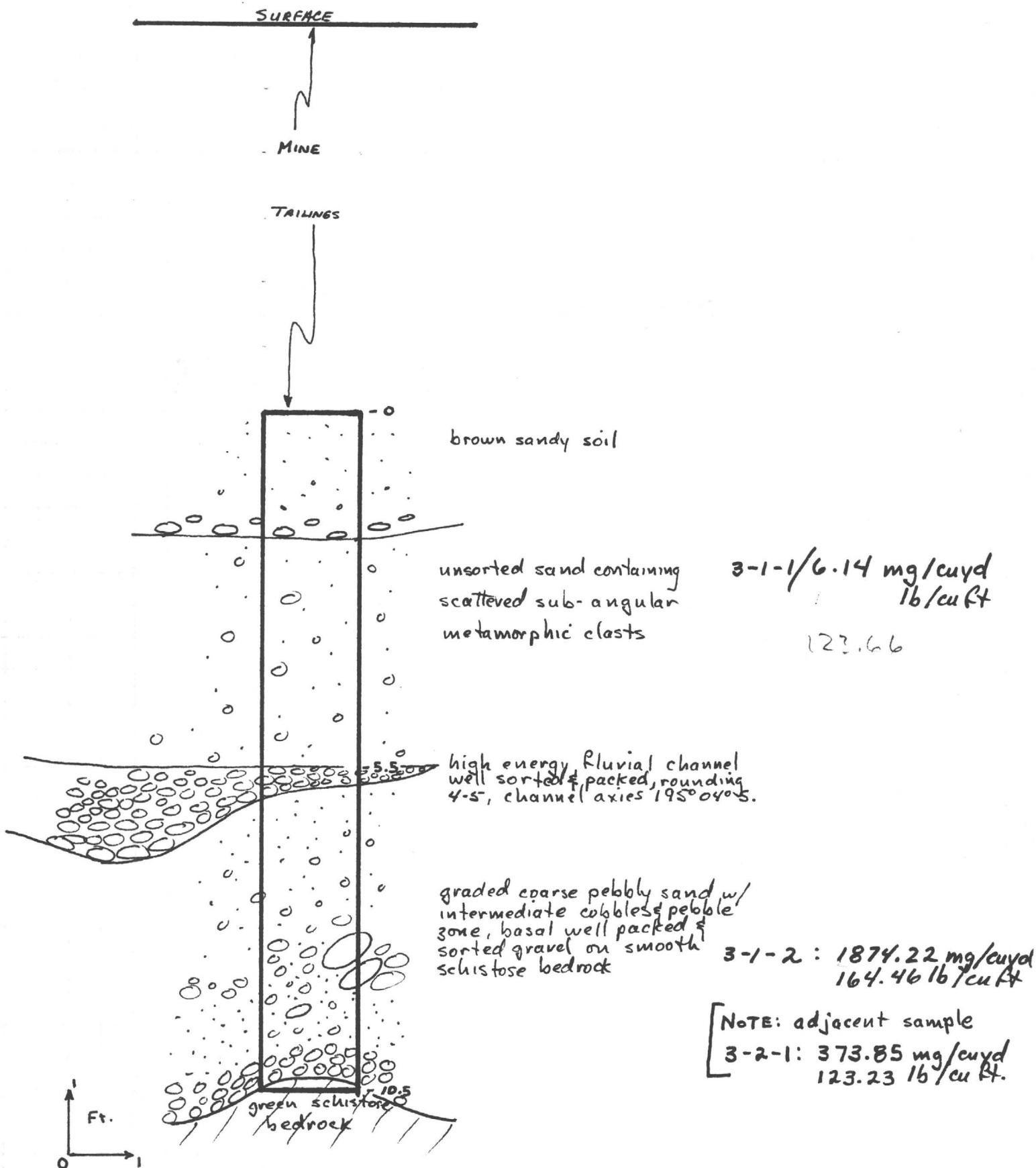
# VULTURE MINE PLACER PROJECT TRENCH 2 CHANNEL 1



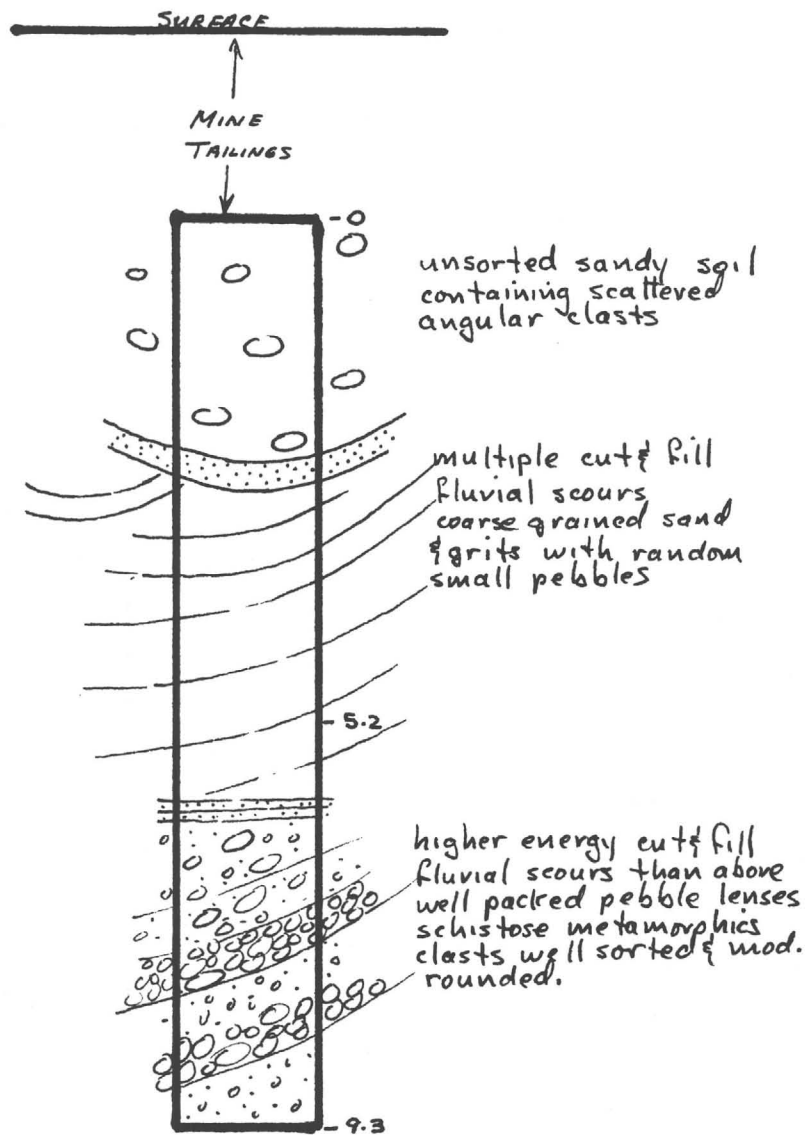
contains lower energy fluvial sedimentation than Trench 1 seen as smaller pebbles & lower angle trough scours



# VULTURE MINE PLACER PROJECT TRENCH 3 CHANNEL 1

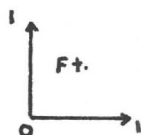
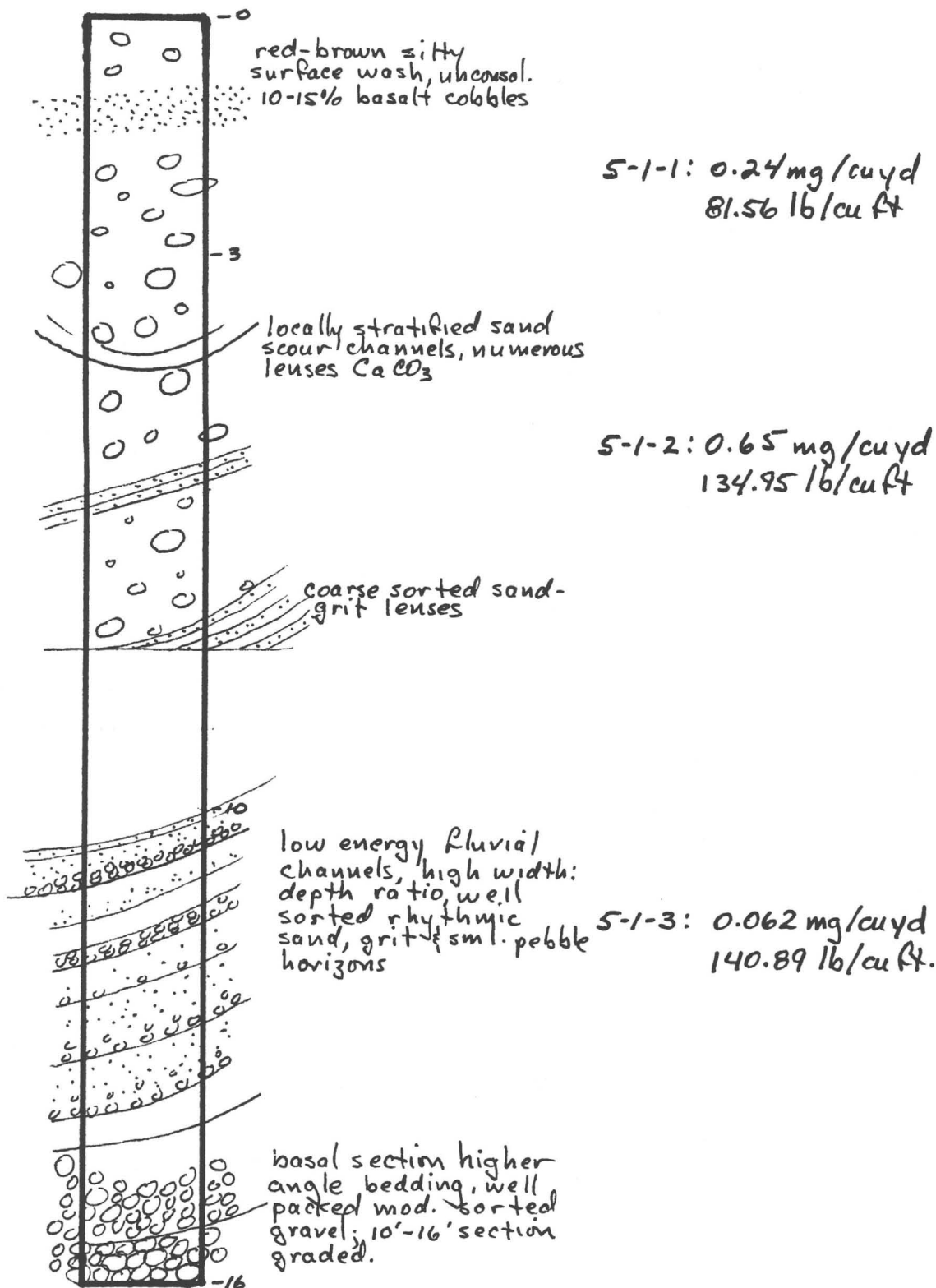


**VULTURE MINE**  
**PLACER PROJECT**  
**TRENCH 4**  
**CHANNEL 1**



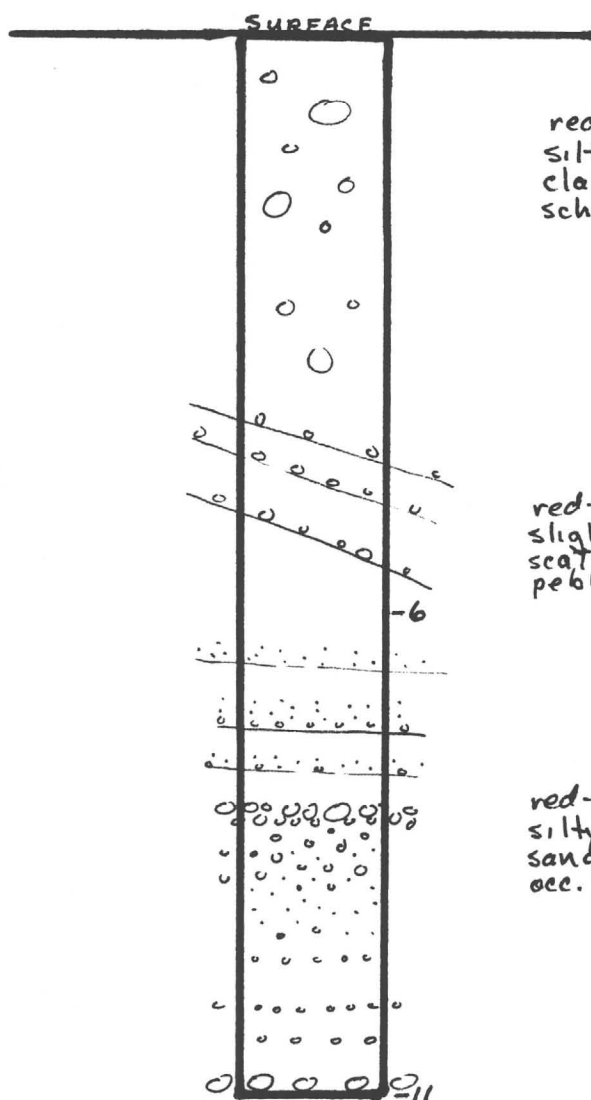
VULTURE MINE  
PLACER PROJECT  
TRENCH 5  
CHANNEL 1

SURFACE





VULTURE MINE  
PLACER PROJECT  
TRENCH 6  
CHANNEL 1



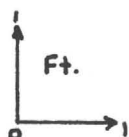
red-brown soil,  
 silt & sand scattered  
 clasts basalt, diorite &  
 schistose metamorphic

6-1-1: 0.42 mg/cuyd  
 93.3 lb/cu ft

red-brown sand  
 slightly winnowed  
 scattered clasts/  
 pebble

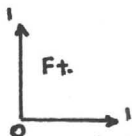
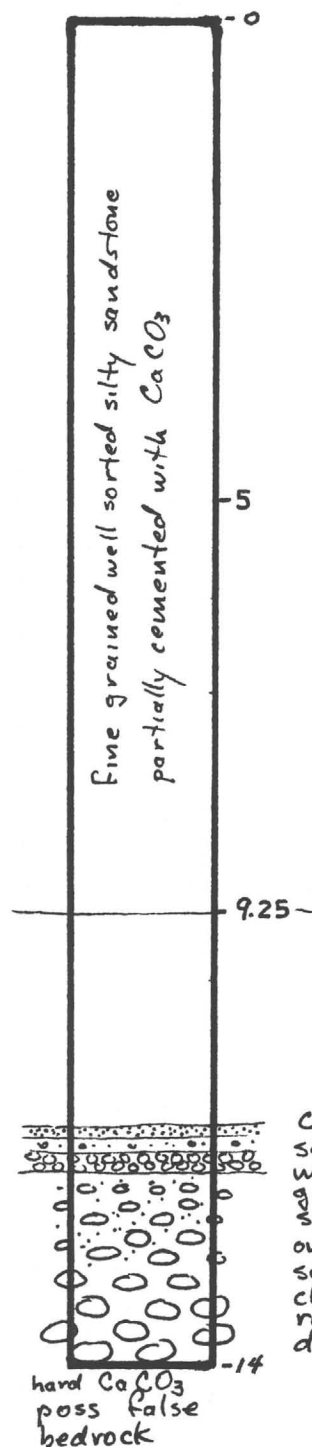
red-brown slightly  
 silty coarse grained  
 sand & grit containing  
 occ. pebble zone

6-1-2: 2.09 mg/cuyd  
 116.13 lb/cu ft.



VULTURE MINE  
PLACER PROJECT  
TRENCH 7  
CHANNEL 1

SURFACE

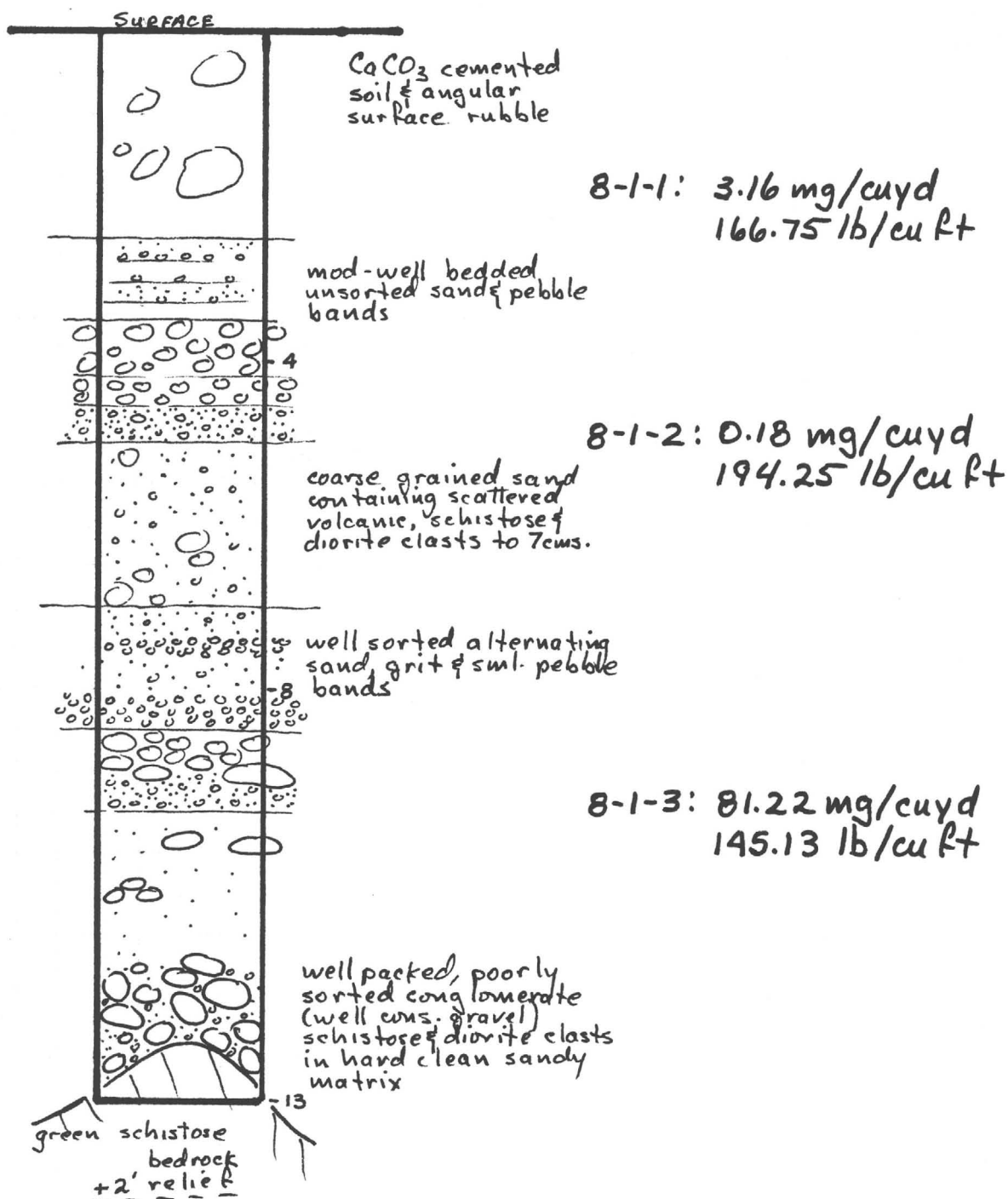


# VULTURE MINE

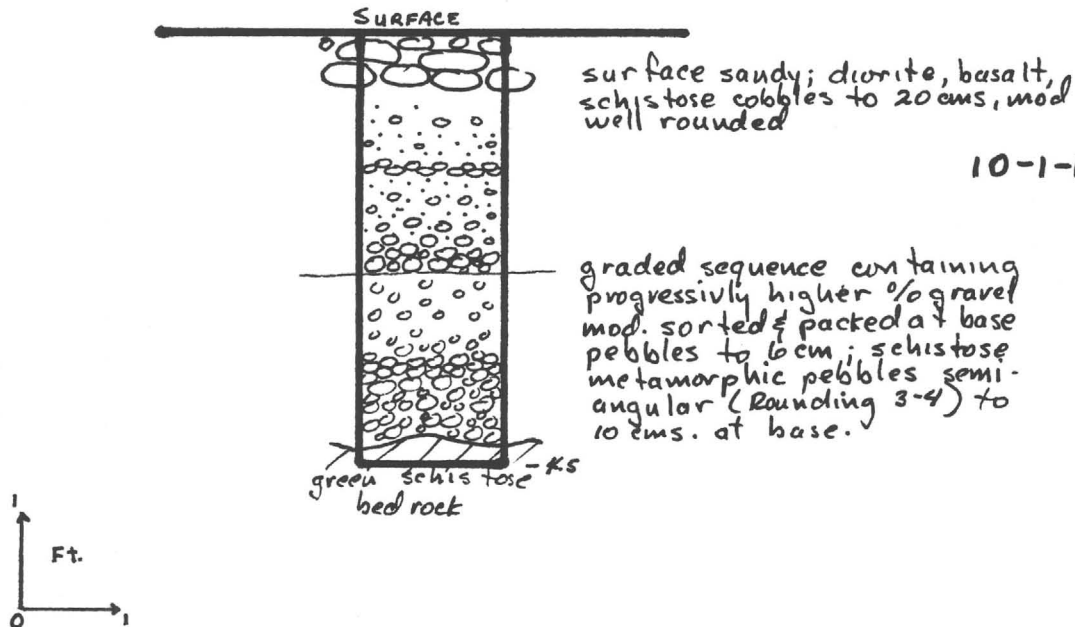
## PLACER PROJECT

### TRENCH 8

### CHANNEL 1



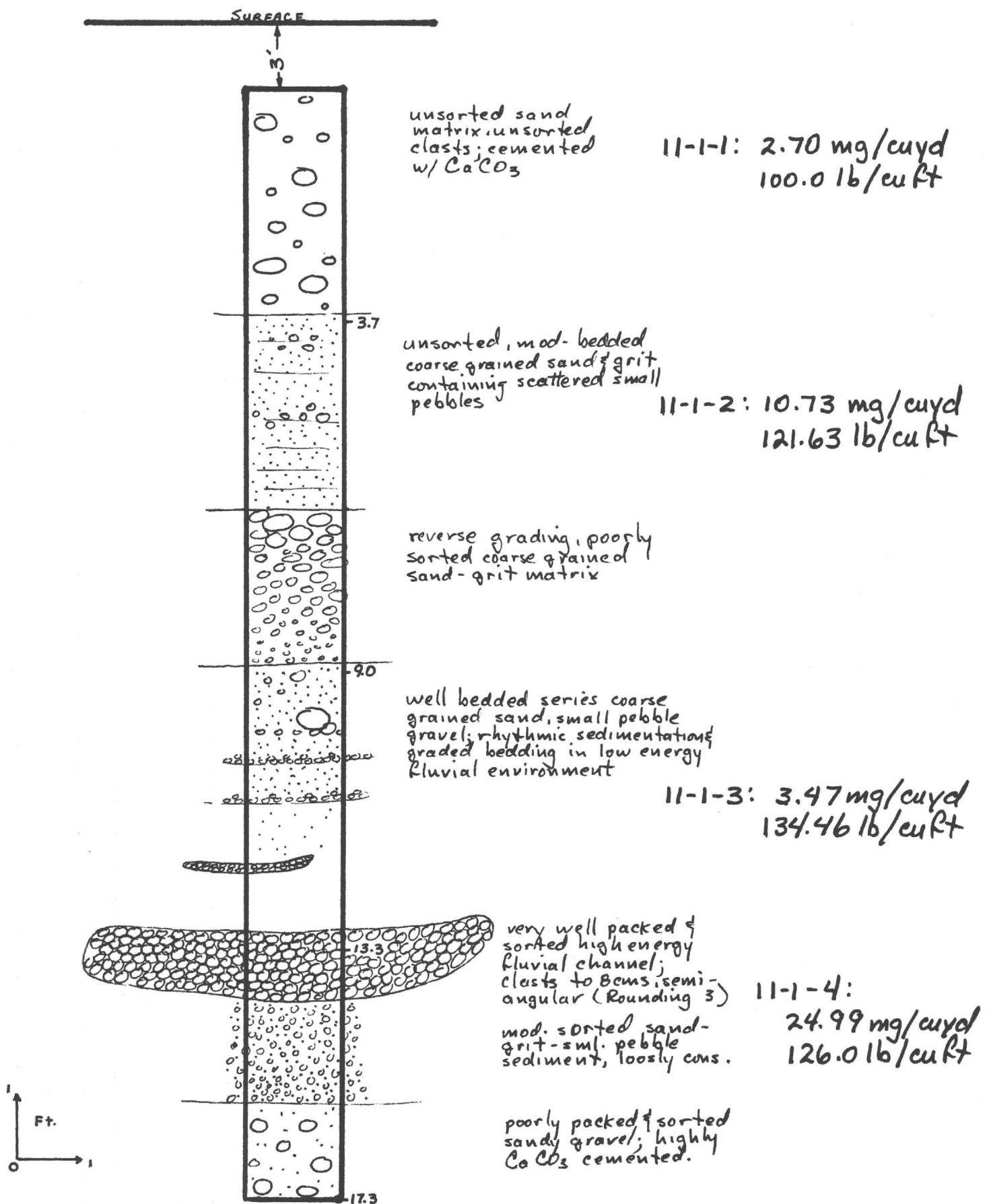
# VULTURE MINE PLACER PROJECT TRENCH 10 CHANNEL 1



10-1-1: 45.92 mg/cuyd  
141.08 lb/cuft



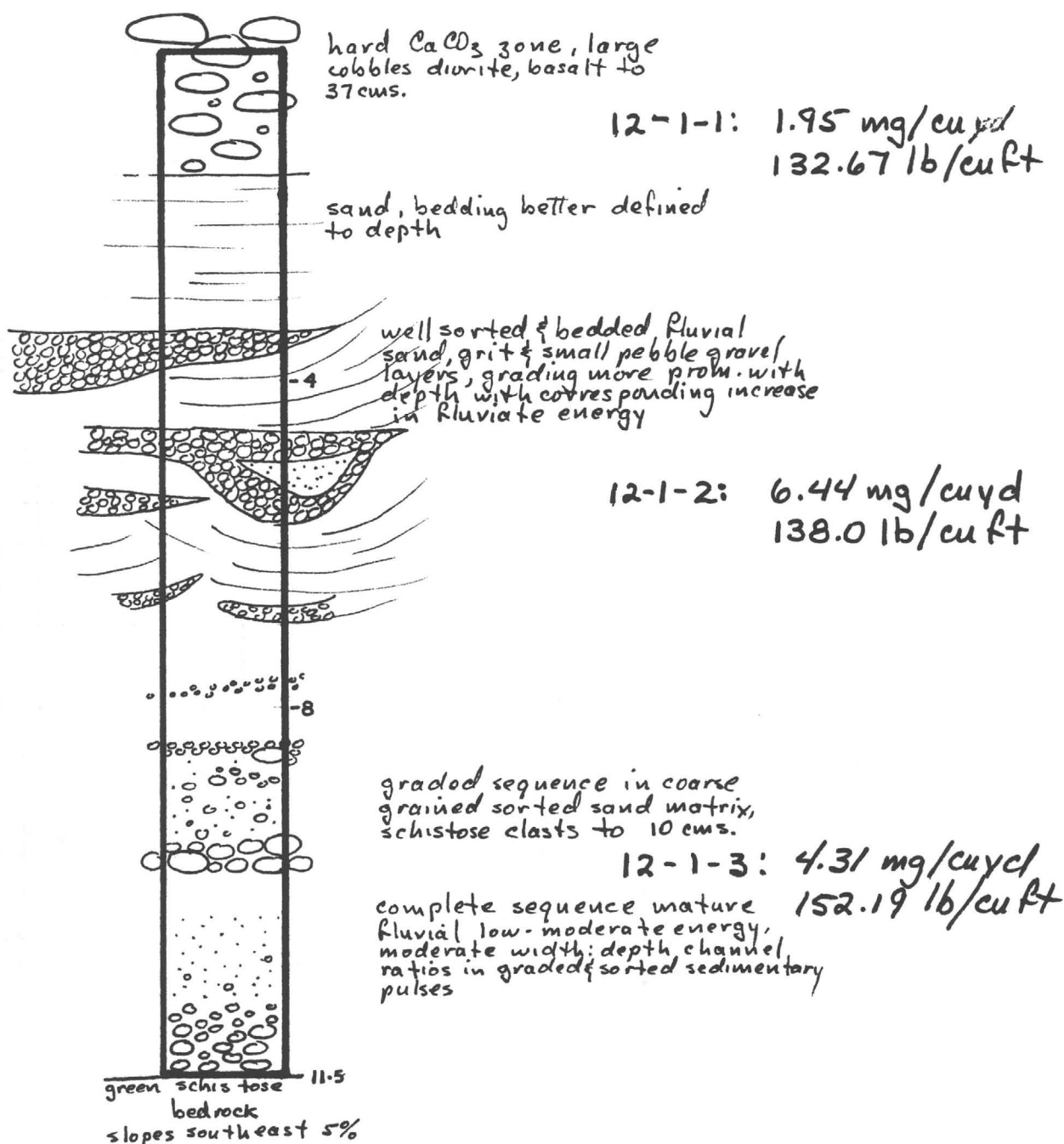
# VULTURE MINE PLACER PROJECT TRENCH 11 CHANNEL 1



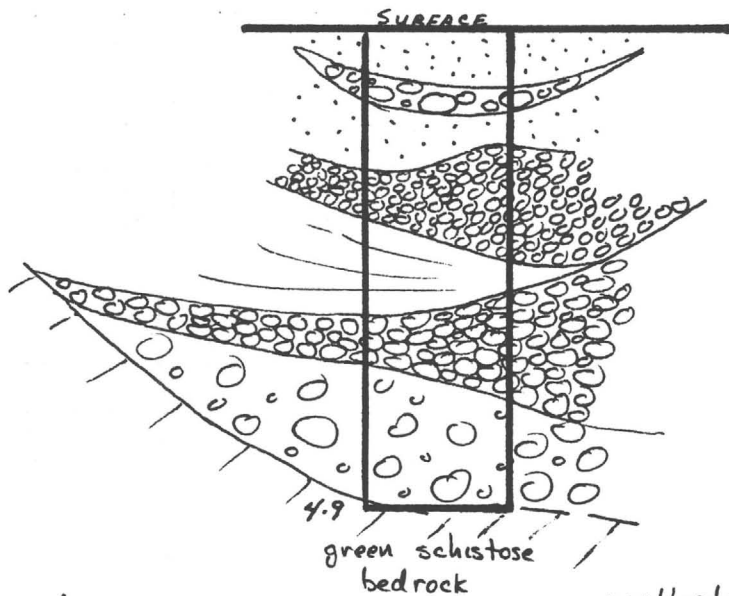
# VULTURE MINE PLACER PROJECT TRENCH 12 CHANNEL 1

SURFACE

surface red-brown  
soil, scattered cobbles  
in  $\text{CaCO}_3$  cemented  
sandy-loam matrix



# VULTURE MINE PLACER PROJECT TRENCH 13 CHANNEL 1



minor  $\text{CaCO}_3$  cement  
in sand matrix

13-1-1:

well packed & sorted  
small pebble gravel, high  
energy, scour channel,  
clasts semi-angular (Rounding 3-4)

375.69 mg/cuy

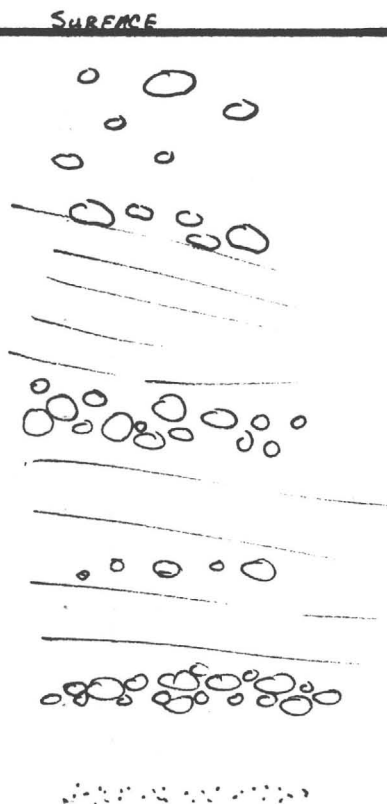
118.58 lb/cu ft

moderately sorted, well  
packed small pebble  
gravel (2cm. avg.)

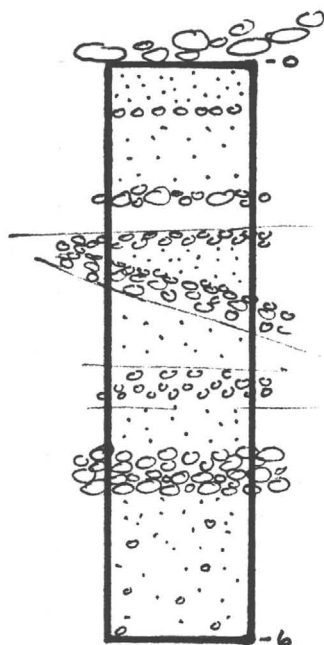
poorly packed & sorted  
coarse sand containing  
10cm schistose clasts

multiple fluvial pulses high  
energy and lap onto well  
incised bedrock channel contour

VULTURE MINE  
PLACER PROJECT  
TRENCH 14  
CHANNEL 1



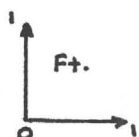
Sequence of poorly bedded & sorted coarse grained sand, grit & pebbly sand units



cyclic fluvial low-moderate energy sed. pulses in wide channel environment

14-1-1: 7.37 mg/cuyd  
 114.0 Yb/cu ft

gravel lenses mod. packed & sorted, clasts semi-angular (Rounding 3) in gritty sand matrix; sequence largely unconsolidated.



VULTURE MINE  
PLACER PROJECT  
TRENCH 15  
CHANNEL 1

SURFACE

fine grained sands  
 containing lenses grits &  
 small pebbles; highly  
 cemented with  $CaCO_3$

unsorted unconsolidated  
 weak bedded cobble unit;  
 schistose metamorphics to  
 15cms. semi-angular (Rounding 3)

zone of rhythmic graded  
 low energy fluvial sediments,  
 basal gravel mod-well packed  
 diorite (Rounding 5) schistose  
 metamorphic (Rounding 4)

15-1-1:

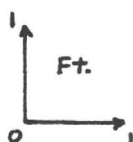
0.88 mg/cuyd  
 112.73 lb/cuft

moderately sorted coarse  
 grained sand-grit

15-1-2:

59.4 mg/cuyd  
 152.31 lb/cuft.

cobble-pebble clasts  
 to 15cms, semi-rounded  
 (Rounding 5-6) in coarse  
 grained sand matrix  
 suggesting low-moderate  
 level fluvial energy



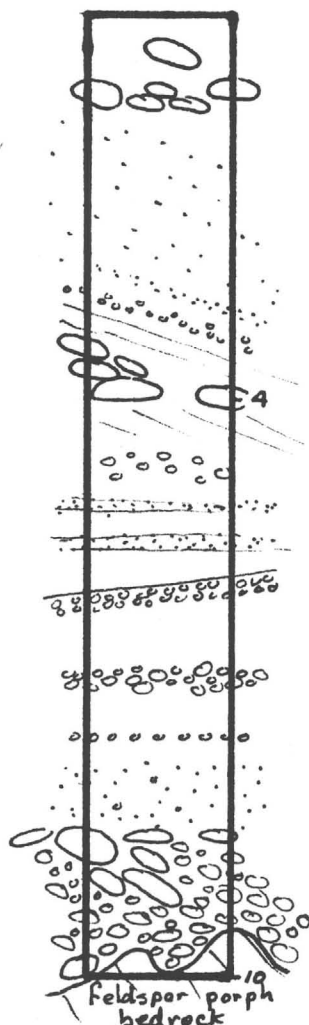


VULTURE MINE  
PLACER PROJECT  
TRENCH 16  
CHANNEL 1

SURFACE

redish sandy soil

highly  $\text{CaCO}_3$  cemented  
 unsorted coarse grained  
 sand containing highly  
 weathered dioritic elasts  
 to 23 cms.



16-1-1: 8.10 mg/cuyd  
 103.83 lb/cu ft

graded sequence, several  
 fluvial cycles of increasing  
 grain size; alternating coarse  
 grained sand bodies with grit  
 and small pebble lenses  
 clasts 100% diorite

16-1-2: 222.48 mg/cuyd  
 114.33 lb/cu ft.

basal section unsorted pebble-  
 cobble gravel composed 100%  
 dioritic identical to bedrock;  
 very local source for the semi-  
 angular (Rounding 4) clasts;  
 bedding gradients 11% suggests  
 high energy environment.

APPENDIX II

# APPENDIX II

## INDIVIDUAL NUGGET WEIGHTS

Sample No.	Mg. Weight	Size	Shape
3-1-2	277.0	5 mm	Elptical spongy
		2.5 mm	Elptical spongy
	Total-----	1.5 mm	"x" moderately round
3-2-1	79.05	4x2 mm	Sub-rounded, spongy
	6.99		
	4.297		
13-1-1	37.06 mg	4x3 mm	Irregular spongy & wire
	19.326 mg	1x4 mm	Wire-Irregular
	4.756		#1 Color equal
			dimensional & flat
	2.306		#1 color equal
			dimensional and flat
	1.960		#1 color equal
			dimensional and flat

APPENDIX III

**JACOBS ASSAY OFFICE****REGISTERED ASSAYERS**

PHONE 622-0813

1435 SOUTH 10TH AVE.

TUCSON, ARIZONA 85713

James M. Prudden  
Consulting Geologist  
4809 Quail Point Rd.  
Salt Lake City, Utah 84124

1/11/85

"Sample Processing Procedures"  
DMEA PLACER PROJECT

Dear Jim:

As-per to our phone conversation yesterday concerning our procedures to our processing the DMEA'S placer project.

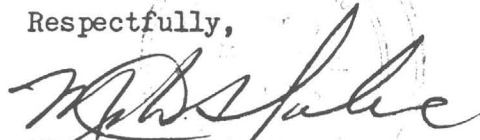
(1) Blacks and Concentrates, will be dried, weighed and submitted to a 4 hour ball mill type grind, any visable gold will first be extracted and weighed prior to milling. An equal amount of water will be added and a slight amount of conc. nitric acid, to help remove any Fe, Ca, and native Hg amalgam (pH checked). After 4 hours have passed, sample is removed and Sodium Hydroxide is added to neturalize pH. Sample is then bottle rolled with Hg (1 gram Hg to 5 grams ore) at approx. 30 RPM for a period of 8 hours. Amalgam is then removed, and the residue is dried and Fire Assayed for Gold & Silver reported Au & Ag oz./ton. Amalgamated material is then subjected to a nitric acid digestion until dissolved, remaining gold is cupelled, weighed and saved.

(2) Black sand concentrates whn designated will, be first dried, weighed, and the entire sample fire assayed for gold & silver. Results reported in total mg. of gold and total mg of silver.

(3) Tailing samples whrn designated, will be fire assayed, using a two (2) assay ton charge (58.33 grams). Results will be reported in troy oz/2000 lb.ton.

We hope this procedure meets with your approval.

Respectfully,



MICHAEL G. JACOBS  
REGISTERED ASSAYER-OWNER



JACOBS ASSAY OFFICE  
1435 So. 10<sup>th</sup> Avenue  
Tucson, Arizona 85713  
602-622-0813

DMEA LTD.

PRELIMINARY TESTS

SAMPLE NO. 1/1/0-6 conc as received.  
Weight as received 927 gr. ~~lb~~; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.217 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

Iron  
mg

METALLICS

PULP

GOLD oz/T XXXXXXX

0.005

Wt. g           

SILVER oz/T XXXXXXX

<0.05

Oz/T gold           

TOTAL mg Gold 6.734

0.1589

Oz/T

TOTAL mg

silver           

silver XXX

N/A

Wt. Gold

Wt. Gold

mg           

mg           

Wt. silver

Wt. silver

mg           

mg           

Percent recovery

Free sodium cyanide  
lb/Ton

gold = 97.69

Solution =           

silver = N/A

Ore =           

927/29.166 = 31.78 A.T.

6.8929/31.78 = .2169 OZ/TON Au

Lime lb/Ton

Solution =           

927 grams milled 4 hours

8 hours agitated w/195 gr. Hg.

Ore =           

Percent recovery

gold =           

silver =           

pH           

Registered Assayer  
Michael G. Jacobs

JACOBS ASSAY OFFICE  
1435 So. 10<sup>th</sup> Avenue  
Tucson, Arizona 85713  
602-622-0813

DMEA LTD.

PRELIMINARY TESTS

SAMPLE NO. 1/1/6-10 conc. as received.  
Weight as received 673 gr. lbs; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.127 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION

200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

TAILS

CYANIDATION

200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

METALLICS

200 grams - 100 Mesh

10 minute screening  
time

→ Iron  
mg

METALLICS

PULP

GOLD oz/T XXXXXXX

0.007

SILVER oz/T XXXXXXX

0.05

TOTAL mg Gold 2.780

0.161

TOTAL mg  
silver N/A

1.15

Wt. g           

Oz/T gold           

Oz/T

silver           

Wt. Gold           

Wt. Gold

mg           

mg           

Percent recovery

gold = 94.53

silver = N/A

Free sodium cyanide

1b/Ton

Solution =           

Ore =           

Wt. silver

mg           

Wt. silver

mg           

$673/29.166 = 23.07$  A.T.

$2.941/23.07 = .127$

Lime 1b/Ton

Solution =           

Ore =           

673 grams milled 4 hours

8 hours agitated w/135 gr.Hg. Percent recovery

gold =           

silver =           

pH           

*Michael G. Jacobs*

Registered Assayer  
Michael G. Jacobs

JACOBS ASSAY OFFICE  
1435 So. 10<sup>th</sup> Avenue  
Tucson, Arizona 85713  
602-622-0813

DMEA LTD.

PRELIMINARY TESTS

SAMPLE NO. 1/1/10'-14.5" concs received.  
Weight as received 4635 gr. ~~1 lb~~ ; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.085 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 9 hours

AMALGAM

TAILS

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening time

Iron  
mg

METALLICS

PULP

GOLD oz/T XXXXXXX

0.002

SILVER oz/T XXXXXXX

0.05

TOTAL mg Gold 13.145

0.318

TOTAL mg silver N/A

7.95

Wt. g           

Oz/T gold           

Oz/T silver           

Wt. Gold           

mg           

Wt. silver           

mg           

Wt. Gold           

mg           

Wt. silver           

mg           

Percent recovery

gold = 97.64

silver = N/A

Free sodium cyanide  
lb/Ton

Solution =           

Ore =           

Lime lb/Ton

Solution =           

Ore =           

Percent recovery

gold =           

silver =           

PH           

$4635/29.166 = 158.92$

$13.463/158.92 = 0.085$

4635 grams milled 4 hours  
8 hour agitated w/930 gr.Hg.

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602-622-0813

PRELIMINARY TESTS

SAMPLE NO. 1/2/1 (4'-8') as received.  
Weight as received 503 gr. lbs; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.058 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams xxx mg; SILVER/200 grams xxx

AMALGAMATION

200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

TAILS

CYANIDATION

200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

METALLICS

200 grams - 100 Mesh

10 minute screening  
time

Iron  
mg

METALLICS

PULP

GOLD oz/T XXXXXXX

0.010

Wt. g           

SILVER oz/T XXXXXXX

0.05

Oz/T gold           

TOTAL mg Gold 0.835

0.173

Oz/T

TOTAL mg

silver           

silver N/A

Wt. Gold           

Wt. Gold

mg           

mg           

Percent recovery

Free sodium cyanide

Wt. silver           

Wt. silver

gold = 82.84

lb/Ton

mg           

mg           

Solution =           

silver = N/A

Ore =           

$503/29.166 = 17.25$

$1.008/17.25 = 0.058$

Lime lb/Ton

Solution =           

503 grams milled 4 hours

8 hours agitated w/100 gr.Hg.

Ore =           

Percent recovery

gold =           

silver =           

pH           

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PRELIMINARY TESTS

SAMPLE NO. 1/2/2 conc. as received.  
Weight as received 399 gr. ~~lb~~; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.249 (calo) oz/T, Silver N/A oz/T  
GOLD/200 grams XXXX mg; SILVER/200 grams XXXX

AMALGAMATION

200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

TAILS

CYANIDATION

200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

METALLICS

200 grams - 100 Mesh

10 minute screening  
time

Iron  
mg

METALLICS

PULP

GOLD oz/T XXXXXXX 0.014  
SILVER oz/T XXXXXXX 0.05  
TOTAL mg Gold 3.210 0.192  
TOTAL mg  
silver N/A 0.68

Percent recovery

gold = \_\_\_\_\_

silver = N/A

$399/29.166 = 13.68$  A.T.

$3.402/13.68 = 0.249$

399 grams milled 4 hours  
8 hours agitated w/80 gr.Hg.

Free sodium cyanide  
lb/Ton  
Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Lime lb/Ton  
Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Percent recovery

gold = \_\_\_\_\_

silver = \_\_\_\_\_

pH \_\_\_\_\_

Wt. g \_\_\_\_\_

Oz/T gold \_\_\_\_\_  
Oz/T

silver \_\_\_\_\_  
Wt. Gold \_\_\_\_\_ Wt. Gold  
mg \_\_\_\_\_ mg

Wt. silver \_\_\_\_\_ Wt. silver  
mg \_\_\_\_\_ mg

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PRELIMINARY TESTS

SAMPLE NO. 2/1/1 conc. as received.  
Weight as received 722 gr. ~~lbs~~; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.033 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

↓  
AMALGAM

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

↓  
SOLUTION

↓  
TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

↓  
Iron

mg

↓  
METALLICS

↓  
PULP

GOLD oz/T XXXXXXX 0.002  
SILVER oz/T XXXXXXX 0.05  
TOTAL mg Gold 0.765 0.050  
TOTAL mg  
silver N/A 1.24

Percent recovery

gold = \_\_\_\_\_

silver = \_\_\_\_\_

$722/29.166 = 24.75$

$.815/24.75 = .033$

722 grams milled 4 hours

8 hours agitated w/145 grams Hg.

Free sodium cyanide  
1b/Ton  
Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Lime 1b/Ton

Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Percent recovery

gold = \_\_\_\_\_

silver = \_\_\_\_\_

pH \_\_\_\_\_

Wt. g \_\_\_\_\_

Oz/T gold \_\_\_\_\_

Oz/T \_\_\_\_\_

silver \_\_\_\_\_

Wt. Gold \_\_\_\_\_ Wt. Gold

mg \_\_\_\_\_ mg \_\_\_\_\_

Wt. silver \_\_\_\_\_ Wt. silver

mg \_\_\_\_\_ mg \_\_\_\_\_

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PRELIMINARY TESTS

SAMPLE NO. 2/1/2 conc. as received.  
Weight as received 1126 gr. lbs ; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.243 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams xxx mg; SILVER/200 grams xxx

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

↓  
AMALGAM

↓  
TAILS

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

↓  
SOLUTION

↓  
TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

↓  
Iron  
mg

↓  
METALLICS

↓  
PULP

GOLD oz/T XXXXXXX 0.004  
SILVER oz/T XXXXXXX 0.05  
TOTAL mg Gold 9.22 0.154  
TOTAL mg  
silver N/A 1.93

Wt. g \_\_\_\_\_  
Oz/T gold \_\_\_\_\_  
Oz/T  
silver \_\_\_\_\_  
Wt. Gold \_\_\_\_\_ Wt. Gold  
mg \_\_\_\_\_ mg

Wt. silver \_\_\_\_\_ Wt. silver  
mg \_\_\_\_\_ mg

Percent recovery

gold = 98.36

silver = N/A

Free sodium cyanide  
lb/Ton

Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Lime lb/Ton

Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Percent recovery

gold = \_\_\_\_\_

silver = \_\_\_\_\_

pH \_\_\_\_\_

$1126/29.166 = 38.61$  A.T.

$9.374/38.61 = 0.243$

1126 grams milled 4 hours

8 hours agitated w/225 gr.Hg.

  
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PRELIMINARY TESTS

SAMPLE NO. 3/1/1 conc as received.  
Weight as received 1643 gr. lbs ; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.048 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium cyanide  
1,2 grams lime  
Agitate 24 hours

METALLICS  
200 grams - 100 Mesh

10 minute screening time

AMALGAM

TAILS

SOLUTION

TAILS

METALLICS

PULP

GOLD oz/T XXXXXXXX

0.013

Wt. g           

SILVER oz/T XXXXXXXX

<0.05

Oz/T gold           

TOTAL mg Gold 1.990

0.732

Oz/T silver           

TOTAL mg silver N/A

N/A

Wt. Gold mg           

Wt. Gold mg           

Wt. silver mg           

Wt. silver mg           

Percent recovery

Free sodium cyanide  
lb/Ton

Wt. silver mg           

Wt. silver mg           

gold = 73.11

Solution =           

silver = N/A

Ore =           

$1643/29.166 = 56.33$

$2.722/56.33 = 0.048$

1643 grams milled 4 hours

8 hours agitated w/330 gr.Hg.

Lime lb/Ton

Solution =           

Ore =           

Percent recovery

gold =           

silver =           

pH           

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PRELIMINARY TESTS

SAMPLE NO. 3/1/2 conc as received.  
Weight as received 553 gr. XXX ; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 12.897 <sup>(calc)</sup> oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

TAILS

SOLUTION

TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

Iron  
mg

METALLICS

PULP

GOLD oz/T XXXXXXXX 0.023  
SILVER oz/T XXXXXXXX <0.05  
TOTAL mg Gold 244.10 0.436  
TOTAL mg  
silver N/A N/A

Percent recovery

gold = 99.82

silver = N/A

Free sodium cyanide  
lb/Ton  
Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Lime lb/Ton  
Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Percent recovery

gold = \_\_\_\_\_

silver = \_\_\_\_\_

pH \_\_\_\_\_

Wt. g \_\_\_\_\_

Oz/T gold \_\_\_\_\_

Oz/T silver \_\_\_\_\_

Wt. Gold \_\_\_\_\_ Wt. Gold  
mg \_\_\_\_\_ mg

Wt. silver \_\_\_\_\_ Wt. silver  
mg \_\_\_\_\_ mg

553/29.166 = 18.96 A.T.  
244.536/18.96 = 12.897 OZ/TON Au

553 grams milled 4 hours  
8 hours agitated w/110 gr.Hg.

Note:

207.0 Mg free gold weighed  
prior to amalgamating.

244.10  
207.00

451.10

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## PRELIMINARY TESTS

SAMPLE NO. 3/2/1 conc. as received.  
Weight as received 612 gr. ~~xxx~~; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 5.15 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

Iron  
mg

METALLICS

PULP

GOLD oz/T XXXXXXX0.033SILVER oz/T XXXXXXX0.05TOTAL mg Gold 107.310.692TOTAL mg  
silverN/A1.05Wt. g           Oz/T gold           

Oz/T

silver           Wt. Gold           mg           Wt. Gold           mg           

Percent recovery

gold = 99.36silver = N/AFree sodium cyanide  
lb/TonSolution =           Ore =           

Lime lb/Ton

Solution =           Ore =           Wt. silver  
mgWt. silver  
mg $612/29.166 = 20.98$  A.T. $108.00/20.98 = 5.15$ 

612 grams milled 4 hours

8 hours agitated w/125 gr.Hg.

Note:

Mg. of free gold wts. 79.056.7994.29790.146gold =           silver =           pH           

Percent recovery

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PRELIMINARY TESTS

SAMPLE NO. 4/1/1 none as received.  
Weight as received 424 gr. lbs ; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.279 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams xxx mg; SILVER/200 grams xxx

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

TAILS

GOLD oz/T XXXXXXX

0.045

SILVER oz/T XXXXXXX

<0.05

TOTAL mg Gold 3.410

0.654

TOTAL mg  
silver N/A

N/A

Percent recovery

gold = 83.91

silver = N/A

$424/29.166 = 14.54$

$4.064/14.54 = .279$  OZ/TON Au

429 grams milled 4 hours

8 hours agitated w/85 gr.Hg.

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

Free sodium cyanide  
lb/Ton

Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Lime lb/Ton

Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Percent recovery

gold = \_\_\_\_\_

silver = \_\_\_\_\_

PH \_\_\_\_\_

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

Iron  
mg

METALLICS

PULP

Wt. g \_\_\_\_\_

Oz/T gold \_\_\_\_\_

Oz/T

silver \_\_\_\_\_

Wt. Gold \_\_\_\_\_

mg \_\_\_\_\_

Wt. silver \_\_\_\_\_

mg \_\_\_\_\_

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PRELIMINARY TESTS

SAMPLE NO. 4/1/2 conc. as received.  
Weight as received 413 gr. lbs; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 1.438 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams xxx mg; SILVER/200 grams xxx

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

Iron

mg

METALLICS

PULP

GOLD oz/T XXXXXXX 0.013

SILVER oz/T XXXXXXX <0.05

TOTAL mg Gold 20.180 0.184

TOTAL mg silver N/A N/A

Percent recovery

gold = 99.09

silver = N/A

$413/29.166 = 14.16$  A.T.

$20.364/14.16 = 1.438$  OZ/TON Au

413 grams milled 4 hours

8 hours agitated w/85 gr.Hg.

Free sodium cyanide  
lb/Ton

Solution =           

Ore =           

Lime lb/Ton

Solution =           

Ore =           

Percent recovery

gold =           

silver =           

PH           

Wt. g           

Oz/T gold           

Oz/T silver           

Wt. Gold            Wt. Gold           

mg            mg           

Wt. silver            Wt. silver           

mg            mg           

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## PRELIMINARY TESTS

SAMPLE NO. 4/2/1 conc. as received.  
 Weight as received 468 gr. ~~XXX~~; Size, all minus 3/8 inches  
 HEAD ASSAY: Gold 0.795 (calo) oz/T, Silver N/A oz/T  
 GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
 40 grams mercury  
 1 gram sodium hydroxide  
 Agitate 3 hours

↓  
AMALGAMCYANIDATION  
200 grams - 100 Mesh

800 cc water  
 2 grams sodium  
                   cyanide  
 1,2 grams lime  
 Agitate 24 hours

↓  
SOLUTION↓  
TAILSMETALLICS  
200 grams - 100 Mesh

10 minute screening  
 time

↓  
 ↓  
 Iron  
                   mg

↓  
METALLICS↓  
PULPGOLD oz/T XXXXXXXX0.050Wt. g           SILVER oz/T XXXXXXXX<.05Oz/T gold           TOTAL mg Gold 11.950.803

Oz/T

TOTAL mg

silver           silver N/AN/A

Wt. Gold

Wt. Gold

mg            mg           

Percent recovery

Free sodium cyanide  
1b/Ton

Wt. silver

Wt. silver

gold = 93.70Solution =           mg            mg           silver = N/AOre =           

468/29.166 = 16.05 A.T.

Lime 1b/Ton

12.753/16.05 = .795 oz/ton

Solution =           Ore =           468 gr. milled 4 hours  
8 hours agitated w/95 gr.Hg.

Percent recovery

gold =           silver =           pH           

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PRELIMINARY TESTS

SAMPLE NO. 5/1/1 conc. as received.  
Weight as received 271 gr. ~~lbs~~ ; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.014 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

↓  
AMALGAM

↓  
TAILS

GOLD oz/T XXXXXXX

0.010

SILVER oz/T XXXXXXX

0.10

TOTAL mg Gold 0.040

0.093

TOTAL mg  
silver N/A

0.93

Percent recovery

gold = 30.08 %

silver = N/A

$271/29.166 = 9.29$

$0.133/9.29 = 0.014$  OZ/TON Au

271 grams milled 4 hours

8 hours agitated w/55 gr.Hg.

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

↓  
SOLUTION

↓  
TAILS

Free sodium cyanide  
1b/Ton  
Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Lime 1b/Ton  
Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Percent recovery

gold = \_\_\_\_\_

silver = \_\_\_\_\_

pH \_\_\_\_\_

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

↓  
Iron  
mg

↓  
METALLICS

↓  
PULP

Wt. g \_\_\_\_\_

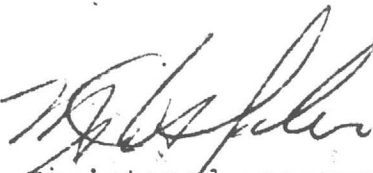
Oz/T gold \_\_\_\_\_

Oz/T

silver \_\_\_\_\_

Wt. Gold \_\_\_\_\_ Wt. Gold  
mg \_\_\_\_\_ mg

Wt. silver \_\_\_\_\_ Wt. silver  
mg \_\_\_\_\_ mg

  
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PRELIMINARY TESTS

SAMPLE NO. 5/1/2 conc. as received.  
Weight as received 780 gr. box; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.013 (calo) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

TAILS

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening time

Iron  
mg

METALLICS

PULP

GOLD oz/T XXXXXXX

.007

Wt. g

SILVER oz/T XXXXXXX

0.10

Oz/T gold

TOTAL mg Gold 0.153

0.187

Oz/T

TOTAL mg

silver N/A

2.67

silver

Wt. Gold

Wt. Gold

mg

mg

Percent recovery

Free sodium cyanide

Wt. silver

Wt. silver

gold = 45.00%

lb/Ton

mg

mg

Solution =

silver = N/A

Ore

=

780/29.166 26.74

Lime lb/Ton

.340/26.74 .013

Solution =

780 grams milled 4 hours

Ore

=

8 hours agitated w/155 gr.Hg.

Percent recovery

gold =

silver =

pH

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## PRELIMINARY TESTS

SAMPLE NO. 5/1/3 conc. as received.  
Weight as received 931 gr. lbs; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.005 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

TAILS

GOLD oz/T XXXXXXX0.003SILVER oz/T XXXXXXX<0.05TOTAL mg Gold 0.0620.096TOTAL mg silver N/AN/A

Percent recovery

gold = 39.24silver = N/A $931/29.166 = 31.92$  $0.158/31.92 = .005 \text{ OZ/TON Au}$ 

931 grams milled 4 hours

8 hours agitated w/185 gr.Hg.

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

Free sodium cyanide  
lb/TonSolution =           Ore =           

Lime lb/Ton

Solution =           Ore =           

Percent recovery

gold =           silver =           pH           

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

→ Iron  
mg

METALLICS

PULP

Wt. g           Oz/T gold           Oz/T silver           Wt. Gold           mg           mg           Wt. silver           mg           

Registered Assayer  
Michael G. Jacobs



JACOBS ASSAY OFFICE  
1435 So. 10<sup>th</sup> Avenue  
Tucson, Arizona 85713  
602-622-0813

DMEA LTD.

PRELIMINARY TESTS

SAMPLE NO. 6/1/1 conc. as received.  
Weight as received 1337 gr. ~~xxx~~ ; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.013<sup>(calc)</sup> oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium cyanide  
1.2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening time

Iron  
mg

METALLICS

PULP

GOLD oz/T XXXXXXX 0.010  
SILVER oz/T XXXXXXX 0.05  
TOTAL mg Gold 0.140 0.458  
TOTAL mg silver N/A 2.292

Wt. g \_\_\_\_\_  
Oz/T gold \_\_\_\_\_  
Oz/T silver \_\_\_\_\_  
Wt. Gold \_\_\_\_\_ Wt. Gold \_\_\_\_\_  
mg \_\_\_\_\_ mg \_\_\_\_\_

Wt. silver \_\_\_\_\_ Wt. silver \_\_\_\_\_  
mg \_\_\_\_\_ mg \_\_\_\_\_

Percent recovery

gold = 23.41

silver = N/A

Free sodium cyanide  
lb/Ton  
Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

$1337/29.166 = 45.84$   
 $.598/45.84 = 0.013 \text{ OZ/TON Au}$

Lime lb/Ton  
Solution = \_\_\_\_\_

1337 grams milled 4 hours  
8 hours agitated w/270 gr.Hg.

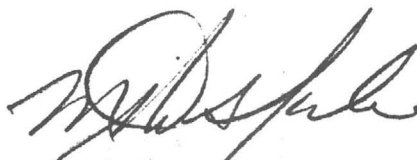
Ore = \_\_\_\_\_

Percent recovery

gold = \_\_\_\_\_

silver = \_\_\_\_\_

PH \_\_\_\_\_

  
Registered Assayer  
Michael G. Jacobs

## PRELIMINARY TESTS

SAMPLE NO. 6/1/2 conc. as received.  
 Weight as received 637 gr. ~~XXX~~; Size, all minus 3/8 inches  
 HEAD ASSAY: Gold 0.038 <sup>(calc)</sup> oz/T, Silver N/A oz/T  
 GOLD/200 grams XXX mg; SILVER/200 grams XXX

## AMALGAMATION

200 grams - 100 Mesh

800 cc water  
 40 grams mercury  
 1 gram sodium hydroxide  
 Agitate 3 hours

AMALGAM

TAILS

## CYANIDATION

200 grams - 100 Mesh

800 cc water  
 2 grams sodium  
                   cyanide  
 1,2 grams lime  
 Agitate 24 hours

SOLUTION

TAILS

## METALLICS

200 grams - 100 Mesh

10 minute screening  
 time

Iron  
 mg

METALLICS

PULP

GOLD oz/T XXXXXXXX0.011Wt. g           SILVER oz/T XXXXXXXX0.05Oz/T gold           TOTAL mg Gold 0.5800.240

Oz/T

TOTAL mg

silver           silver N/A1.09Wt. Gold           

Wt. Gold

mg           mg           

Percent recovery

Free sodium cyanide  
lb/Ton

Wt. silver

Wt. silver

gold = 70.73Solution =           mg           mg           silver = N/AOre =            $637/29.166 = 21.84$  A.T.

Lime lb/Ton

 $0.820/21.84 = 0.038$  oz/tonSolution =           Ore =           

637 gr. milled 4 hours

8 hours agitated w/130 gr.Hg.

Percent recovery

gold =           silver =           pH           

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 Registered Assayer  
 Michael G. Jacobs

## PRELIMINARY TESTS

SAMPLE NO. 7/1/1 conc. as received.  
 Weight as received 70 grams ~~XXX~~; Size, all minus 3/8 inches  
 HEAD ASSAY: Gold 0.540 (calc) oz/T, Silver N/A oz/T  
 GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

↓  
AMALGAM

↓  
TAILS

GOLD oz/T XXXXXXX 0.528  
 SILVER oz/T XXXXXXX 0.05  
 TOTAL mg Gold 0.030 1.267  
 TOTAL mg silver N/A N/A

Percent recovery

gold = 2.31

silver = N/A

$70/29.166 = 2.40$  A.T.  
 $1.297/2.40 = 0.540$  oz/ton

70 grams milled 4 hours  
8 hours agitated w/15 gr.Hg.

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium cyanide  
1,2 grams lime  
Agitate 24 hours

↓

SOLUTION

↓

TAILS

Free sodium cyanide  
lb/Ton  
Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Lime lb/Ton  
Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Percent recovery

gold = \_\_\_\_\_

silver = \_\_\_\_\_

pH \_\_\_\_\_

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

↓

METALLICS

→ Iron

mg

↓

PULP

Wt. g \_\_\_\_\_

Oz/T gold \_\_\_\_\_


Oz/T silver \_\_\_\_\_

Wt. Gold \_\_\_\_\_ Wt. Gold \_\_\_\_\_

mg \_\_\_\_\_ mg \_\_\_\_\_

Wt. silver \_\_\_\_\_ Wt. silver \_\_\_\_\_

mg \_\_\_\_\_ mg \_\_\_\_\_

  
 Registered Assayer  
 Michael G. Jacobs

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PRELIMINARY TESTS

SAMPLE NO. 7/1/2 5'-9'-3" concas received.  
Weight as received 86 grams ~~lbs~~ ; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.117 (cal) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

↓  
AMALGAM

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium cyanide  
1,2 grams lime  
Agitate 24 hours

↓  
SOLUTION

↓  
TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening time

↓  
Iron  
mg

↓  
METALLICS

↓  
PULP

GOLD oz/T XXXXXXX 0.096  
SILVER oz/T XXXXXXX 0.05  
TOTAL mg Gold 0.063 0.283  
TOTAL mg silver N/A N/A

Wt. g \_\_\_\_\_

Oz/T gold \_\_\_\_\_

Oz/T silver \_\_\_\_\_

Wt. Gold \_\_\_\_\_ Wt. Gold \_\_\_\_\_

mg \_\_\_\_\_ mg \_\_\_\_\_

Wt. silver \_\_\_\_\_ Wt. silver \_\_\_\_\_

mg \_\_\_\_\_ mg \_\_\_\_\_

Percent recovery

gold = 11.73

silver = N/A

Free sodium cyanide  
lb/Ton

Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Lime lb/Ton

Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Percent recovery

gold = \_\_\_\_\_

silver = \_\_\_\_\_

pH \_\_\_\_\_

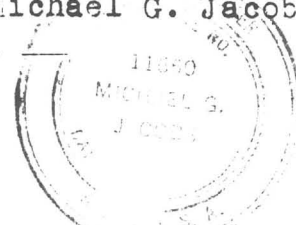
86/29.166 = 2.949 A.T.

0.346/2.949 = 0.117 oz/ton

86 grams milled 4 hours

8 hours agitated w/20 gr.Hg.

*Michael G. Jacobs*  
Registered Assayer  
Michael G. Jacobs



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PRELIMINARY TESTS

SAMPLE NO. 7/1/3 9'-3"-14'-3"<sup>conc</sup> as received.  
Weight as received 949 grams ~~XXX~~; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.006 (cal) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

↓  
AMALGAM

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

↓  
SOLUTION

↓  
TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

↓  
Iron  
mg

↓  
METALLICS

↓  
PULP

GOLD oz/T XXXXXXX 0.005  
SILVER oz/T XXXXXXX 0.05  
TOTAL mg Gold 0.022 0.163  
TOTAL mg  
silver N/A N/A

Percent recovery

gold = 11.89

silver = N/A

$949/29.166 = 32.53$  A.T.  
 $0.185/32.53 = 0.006$  oz/ton  
949 grams milled 4 hours  
8 hours agitated w/190 gr.Hg.

Free sodium cyanide  
lb/Ton  
Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Lime lb/Ton  
Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Percent recovery

gold = \_\_\_\_\_

silver = \_\_\_\_\_

pH \_\_\_\_\_

Wt. g \_\_\_\_\_

Oz/T gold \_\_\_\_\_

Oz/T \_\_\_\_\_


silver \_\_\_\_\_

Wt. Gold \_\_\_\_\_ Wt. Gold \_\_\_\_\_

mg \_\_\_\_\_ mg \_\_\_\_\_

Wt. silver \_\_\_\_\_ Wt. silver \_\_\_\_\_

mg \_\_\_\_\_ mg \_\_\_\_\_

  
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PRELIMINARY TESTS

SAMPLE NO. 8/1/1 0-4' conc. as received.  
Weight as received 349 grams ~~XXX~~; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.008 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening time

Iron

mg

METALLICS

PULP

GOLD oz/T XXXXXXX

0.006

Wt. g           

SILVER oz/T XXXXXXX

<.05

Oz/T gold           

TOTAL mg Gold 0.026

0.072

Oz/T

TOTAL mg

silver           

silver N/A

N/A

Wt. Gold

Wt. Gold

mg           

mg           

Percent recovery

Free sodium cyanide  
lb/Ton

Wt. silver

Wt. silver

gold = 26.53

Solution =           

mg           

mg           

silver = N/A

Ore =           

$349 \text{ gr}/29.166 = 11.97 \text{ A.T.}$

$0.098/11.97 = 0.008 \text{ oz./ton}$

Lime lb/Ton

Solution =           

349 grams milled 4 hours

8 hours agitated w/70 gr.Hg.

Ore =           

Percent recovery

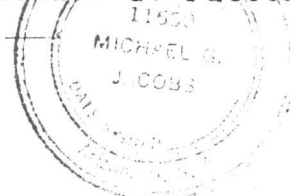
gold =           

silver =           

pH           

*Michael G. Jacobs*

Registered Assayer  
Michael G. Jacobs





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PRELIMINARY TESTS

SAMPLE NO. 8/1/2 4'-8' conc. as received.  
Weight as received 755 grams ~~XXX~~; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.024 (cal) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

Iron  
mg

METALLICS

PULP

GOLD oz/T XXXXXXX

0.006

SILVER oz/T XXXXXXX

<.05

TOTAL mg Gold 0.468

0.155

TOTAL mg

silver N/A

N/A

Wt. g           

Oz/T gold           

Oz/T

silver           

Wt. Gold           

Wt. Gold

mg           

mg           

Percent recovery

gold = 75.12

silver = N/A

Free sodium cyanide  
lb/Ton

Solution =           

Ore =           

Wt. silver

mg           

Wt. silver

mg           

755/29.166 = 25.89 A.T.

0.623/25.89 = 0.024 oz/ton

Lime lb/Ton

Solution =           

Ore =           

755 grams milled 4 hours

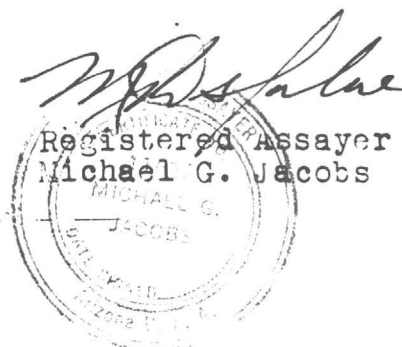
8 hours agitated w/150 gr.Hg.

Percent recovery

gold =           

silver =           

pH           



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PRELIMINARY TESTS

SAMPLE NO. 8/1/3 8'-12'-10"<sup>conc.</sup> as received.  
Weight as received 461 grams ~~XXX~~; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.939<sup>calc.</sup> oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

TAILS

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

Iron

mg

METALLICS

PULP

GOLD oz/T XXXXXXXX

0.020

SILVER oz/T XXXXXXXX

<.05

TOTAL mg Gold 14.530

0.316

TOTAL mg  
silver N/A

N/A

Percent recovery

gold = 97.87

silver = N/A

$461/29.166 = 15.81$  A.T.

$14.846/15.81 = 0.939$  oz/ton

461 grams milled 4 hours

8 hours agitated w/95 gr.Hg.

Free sodium cyanide  
lb/Ton

Solution =           

Ore =           

Lime lb/Ton

Solution =           

Ore =           

Percent recovery

gold =           

silver =           

pH           

Wt. g           

Oz/T gold           

Oz/T           

silver           

Wt. Gold           

mg           

Wt. silver           

mg           

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PRELIMINARY TESTS

SAMPLE NO. 9/1/1 0-4'-4" conc.as received.  
Weight as received 1124 grams ~~XXX~~ ; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.499 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

↓  
AMALGAM

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium cyanide  
1,2 grams lime  
Agitate 24 hours

↓  
SOLUTION

↓  
TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening time

↓  
→ Iron  
mg

↓  
METALLICS

↓  
PULP

GOLD oz/T XXXXXXXX 0.031

SILVER oz/T XXXXXXXX <.05

TOTAL mg Gold 18.020 1.194

TOTAL mg silver N/A N/A

Percent recovery

gold = 93.79

silver = N/A

1124/29.166 = 38.54 A.T.

19.214/38.54 = 0.499 oz/ton

1124 grams milled 4 hours  
8 hours agitated w/225 gr.Hg.

Free sodium cyanide  
lb/Ton  
Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Lime lb/Ton  
Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Percent recovery

gold = \_\_\_\_\_

silver = \_\_\_\_\_

pH \_\_\_\_\_

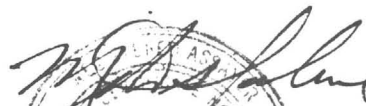
Wt. g \_\_\_\_\_

Oz/T gold \_\_\_\_\_

Oz/T silver \_\_\_\_\_

Wt. Gold \_\_\_\_\_ Wt. Gold  
mg \_\_\_\_\_ mg \_\_\_\_\_

Wt. silver \_\_\_\_\_ Wt. silver  
mg \_\_\_\_\_ mg \_\_\_\_\_

  
Registered Assayer  
Michael G. Jacobs  
JACOBS  
SILVER  
ARIZONA

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PRELIMINARY TESTS

SAMPLE NO. 10/1/1 conc. as received.  
Weight as received 824 grams ~~XXX~~; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.398 (cal) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION

200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

TAILS

CYANIDATION

200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

METALLICS

200 grams - 100 Mesh

10 minute screening  
time

Iron

mg

METALLICS

PULP

GOLD oz/T XXXXXXX

0.014

SILVER oz/T XXXXXXX

<.05

TOTAL mg Gold 11.054

0.396

TOTAL mg  
silver N/A

N/A

Wt. g           

Oz/T gold           

Oz/T

silver           

Wt. Gold            Wt. Gold

mg            mg           

Wt. silver            Wt. silver

mg            mg           

Percent recovery

gold = 96.54

silver = N/A

Free sodium cyanide

lb/Ton

Solution =           

Ore =           

Lime lb/Ton

Solution =           

Ore =           

Percent recovery

gold =           

silver =           

pH           

$824/29.166 = 28.25$  A.T.

$11.45/28.25 = 0.398$  oz/ton

824 grams milled 4 hours

8 hours agitated w/165gr.Hg.

*Michael G. Jacobs*  
Registered Assayer  
Michael G. Jacobs

DMEA LTD.

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602-622-0813

PRELIMINARY TESTS

SAMPLE NO. 11/1/1 conc as received.  
Weight as received 199 grams ~~100~~; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.068 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

↓  
AMALGAM

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium cyanide  
1,2 grams lime  
Agitate 24 hours

↓  
SOLUTION

↓  
TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening time

↓  
Iron  
mg

↓  
METALLICS

↓  
PULP

GOLD oz/T XXXXXXX

0.008

Wt. g           

SILVER oz/T XXXXXXX

0.05

Oz/T gold           

TOTAL mg Gold 0.413

0.054

Oz/T silver           

TOTAL mg

silver N/A

Wt. Gold            Wt. Gold

mg            mg           

Percent recovery

gold = 88.44

silver = N/A

Free sodium cyanide  
lb/Ton

Solution =           

Ore =           

Wt. silver

mg           

Wt. silver

mg           

199/29.166 = 6.82 A.T.

0.467/6.82 = 0.068 oz/ton

Lime lb/Ton

Solution =           

Ore =           

199 grams milled 4 hours

8 hours agitated w/40 gr.Hg.

Percent recovery

gold =           

silver =           

pH           

*Michael G. Jacobs*  
Registered Assayer  
Michael G. Jacobs  
MICHAEL G.  
JACOBS  
CALIFORNIA  
ARIZONA

DMEA LTD.

JACOBS ASSAY OFFICE  
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Tucson, Arizona 85713  
602-622-0813

PRELIMINARY TESTS

SAMPLE NO. 11/1/2 conc. as received.  
Weight as received 784 grams ~~105X~~; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.133 (caloz)/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

TAILS

GOLD oz/T XXXXXXX

0.009

SILVER oz/T XXXXXXX

0.05

TOTAL mg Gold 3.180

0.231

TOTAL mg silver N/A

N/A

Percent recovery

gold = 93.23

silver = N/A

$784/29.166 = 25.65$  A.T.

$3.411/25.65 = 0.133$  oz/ton

784 grams milled 4 hours

8 hours agitated w/155 gr.Hg.

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

Free sodium cyanide  
lb/Ton

Solution =           

Ore =           

Lime lb/Ton

Solution =           

Ore =           

Percent recovery

gold =           

silver =           

pH           

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

Iron  
mg

METALLICS

PULP

Wt. g           

Oz/T gold           

Oz/T silver           

Wt. Gold           

mg           

Wt. Gold           

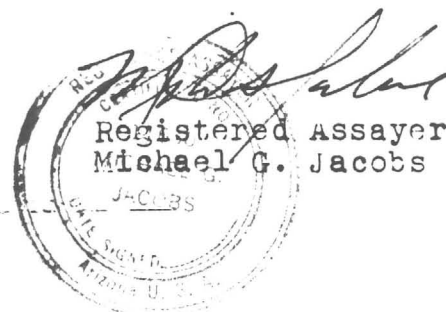
mg           

Wt. silver           

mg           

Wt. silver           

mg           





DMEA LTD.

JACOBS ASSAY OFFICE  
1435 So. 10<sup>th</sup> Avenue  
Tucson, Arizona 85713  
602-622-0813

PRELIMINARY TESTS

SAMPLE NO. 11/1/3 conc. as received.  
Weight as received 867 grams ~~XXX~~; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.030 (cal) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

↓  
AMALGAM

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

↓  
SOLUTION

↓  
TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

↓  
Iron  
mg

↓  
METALLICS

↓  
PULP

GOLD oz/T XXXXXXXX

0.002

SILVER oz/T XXXXXXXX

0.05

TOTAL mg Gold 0.835

0.059

TOTAL mg  
silver N/A

N/A

Percent recovery

gold = 93.40

silver = N/A

$867/29.166 = 29/73$  A.T.

$0.894/29.73 = 0.030$  oz/ton

867 grams milled 4 hours

8 hours agitated w/175 gr.Hg.

Free sodium cyanide  
lb/Ton

Solution =           

Ore =           

Lime lb/Ton

Solution =           

Ore =           

Percent recovery

gold =           

silver =           

pH           

Wt. g           

Oz/T gold           

Oz/T

silver           

Wt. Gold           

mg           

Wt. silver           

mg           

Registered Assayer  
Michael G. Jacobs

DMEA LTD.

JACOBS ASSAY OFFICE  
1435 So. 10<sup>th</sup> Avenue  
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602-622-0813

PRELIMINARY TESTS

SAMPLE NO. 11/1/4 conc. as received.  
Weight as received 385 grams lbs ; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.433 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

TAILS

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening time

Iron  
mg

METALLICS

PULP

GOLD oz/T XXXXXXX

0.012

SILVER oz/T XXXXXXX

0.05

TOTAL mg Gold 5.553

0.158

TOTAL mg silver N/A

N/A

Wt. g           

Oz/T gold           

Oz/T silver           

Wt. Gold            Wt. Gold             
mg            mg           

Wt. silver            Wt. silver             
mg            mg           

Percent recovery

gold = 97.23

silver = N/A

Free sodium cyanide  
lb/Ton  
Solution =           

Ore =           

Lime lb/Ton  
Solution =           

Ore =           

Percent recovery

gold =           

silver =           

pH           

$385/29.166 = 13.20$  A.T.

$5.711/13.20 = 0.433$  oz/ton

385 grams milled 4 hours  
8 hours agitated w/80 gr.Hg.

*Michael G. Jacobs*  
Registered Assayer  
Michael G. Jacobs  
MICHAEL G. JACOBS  
DATE SIGNED  
AZ 2000

JACOBS ASSAY OFFICE  
1435 So. 10<sup>th</sup> Avenue  
Tucson, Arizona 85713  
602-622-0813

DMEA LTD.

PRELIMINARY TESTS

SAMPLE NO. 12/1/1 conc. \_\_\_\_\_ as received.  
Weight as received 587 grams ~~XXX~~ ; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.028 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams xxx mg; SILVER/200 grams xxx

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

TAILS

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening time

Iron  
mg

METALLICS

PULP

GOLD oz/T XXXXXXX

0.006

Wt. g \_\_\_\_\_

SILVER oz/T XXXXXXX

0.05

Oz/T gold \_\_\_\_\_

TOTAL mg Gold 0.434

0.120

Oz/T

TOTAL mg

silver \_\_\_\_\_

silver N/A

N/A

Wt. Gold \_\_\_\_\_

Wt. Gold \_\_\_\_\_

mg \_\_\_\_\_

mg \_\_\_\_\_

Percent recovery

Free sodium cyanide  
lb/Ton

Wt. silver \_\_\_\_\_

Wt. silver \_\_\_\_\_

gold = 78.34

Solution = \_\_\_\_\_

mg \_\_\_\_\_

mg \_\_\_\_\_

silver = N/A

Ore = \_\_\_\_\_

$587/29.166 = 20.13$  A.T.  
 $0.554/20.13 = 0.028$  oz/ton

Lime lb/Ton  
Solution = \_\_\_\_\_

587 grams milled 4 hours  
8 hours agitated w/120gr.Hg.

Ore = \_\_\_\_\_

Percent recovery

gold = \_\_\_\_\_

silver = \_\_\_\_\_

pH \_\_\_\_\_

*Michael G. Jacobs*  
Registered Assayer  
Michael G. Jacobs  
JACOBS  
TUCSON, ARIZONA

DMEA LTD.

JACOBS ASSAY OFFICE  
1435 So. 10<sup>th</sup> Avenue  
Tucson, Arizona 85713  
602-622-0813

PRELIMINARY TESTS

SAMPLE NO. 12/1/2 conc. as received.  
Weight as received 569 grams ~~100~~ XXXX; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.075 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

TAILS

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

Iron  
mg

METALLICS

PULP

GOLD oz/T XXXXXXX 0.002  
SILVER oz/T XXXXXXX 0.05  
TOTAL mg Gold 1.430 0.039  
TOTAL mg silver N/A N/A

Wt. g \_\_\_\_\_  
Oz/T gold \_\_\_\_\_  
Oz/T silver \_\_\_\_\_  
Wt. Gold \_\_\_\_\_ Wt. Gold \_\_\_\_\_  
mg \_\_\_\_\_ mg \_\_\_\_\_

Wt. silver \_\_\_\_\_ Wt. silver \_\_\_\_\_  
mg \_\_\_\_\_ mg \_\_\_\_\_

Percent recovery

gold = 97.35

silver = N/A

Free sodium cyanide  
lb/Ton  
Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

$569/29.166 = 19.51$  A.T.  
 $1.460/19.51 = 0.075$  oz/ton

569 grams milled 4 hours  
8 hours agitated w/115 gr.Hg.

Lime lb/Ton  
Solution = \_\_\_\_\_

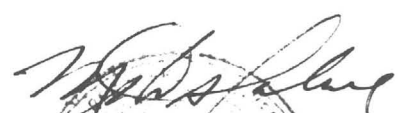
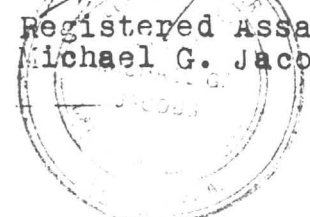
Ore = \_\_\_\_\_

Percent recovery

gold = \_\_\_\_\_

silver = \_\_\_\_\_

pH \_\_\_\_\_

  
Registered Assayer  
Michael G. Jacobs  


DMEA LTD.

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Tucson, Arizona 85713  
602-622-0813

PRELIMINARY TESTS

SAMPLE NO. 12/1/3 conc. as received.  
Weight as received 591 grams ~~XXX~~; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.045 (caloz)/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium cyanide  
1,2 grams lime  
Agitate 24 hours

TAILS

SOLUTION

TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening time

Iron

mg

METALLICS

PULP

GOLD oz/T XXXXXXX

0.004

SILVER oz/T XXXXXXX

0.05

TOTAL mg Gold 0.838

0.081

TOTAL mg silver N/A

N/A

Wt. g           

Oz/T gold           

Oz/T silver           

Wt. Gold            Wt. Gold             
mg            mg           

Wt. silver            Wt. silver             
mg            mg           

Percent recovery

gold = 91.19

silver = N/A

Free sodium cyanide  
lb/Ton  
Solution =           

Ore =           

$591/29.166 = 20/26$  A.T.  
 $0.919/20.26 = 0.045$  oz/ton

Lime lb/Ton  
Solution =           

591 grams milled 4 hours  
8 hours agitated w/120 gr.Hg.

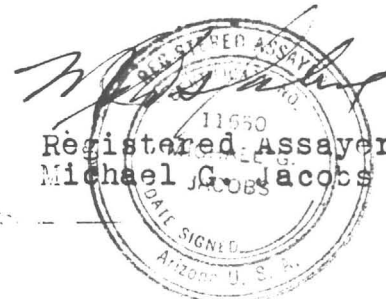
Ore =           

Percent recovery

gold =           

silver =           

pH           



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DMEA LTD.

PRELIMINARY TESTS

SAMPLE NO. 13/1/1 conc. as received.  
Weight as received 289 grams ~~XXX~~; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 7.39 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

TAILS

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

Iron  
mg

METALLICS

PULP

GOLD oz/T XXXXXXX 0.663

SILVER oz/T XXXXXXX <0.05

TOTAL mg Gold 66.650 6.570

TOTAL mg  
silver N/A N/A

Wt. g                      

Oz/T gold                      

Oz/T silver                      

Wt. Gold            Wt. Gold  
mg            mg           

Wt. silver            Wt. silver  
mg            mg           

Percent recovery

gold = 91.02

silver = N/A

Free sodium cyanide  
lb/Ton  
Solution =           

Ore =           

Lime lb/Ton  
Solution =           

Ore =           

Percent recovery

gold =           

silver =           

PH           

Note: Sample wts. of small nuggets.

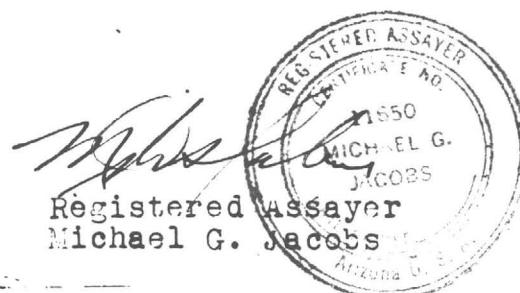
#1--37.060 Mg.

#2--19.326 "

#3--4.756 "

#4--2.306 "

#5--1.960 "





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602-622-0813

DMEA LTD.

PRELIMINARY TESTS

SAMPLE NO. 14/1/1 conc. \_\_\_\_\_ as received.  
Weight as received 1195 grams ~~XXX~~ ; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.063 (caloz)/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

TAILS

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening time

Iron  
mg

METALLICS

PULP

GOLD oz/T XXXXXXXX

0.003

Wt. g \_\_\_\_\_

SILVER oz/T XXXXXXXX

<.05

Oz/T gold \_\_\_\_\_

TOTAL mg Gold 2.455

0.123

Oz/T silver \_\_\_\_\_

TOTAL mg

Wt. Gold \_\_\_\_\_ Wt. Gold

silver N/A

N/A

mg \_\_\_\_\_ mg \_\_\_\_\_

Percent recovery

Free sodium cyanide  
1b/Ton

Wt. silver \_\_\_\_\_ Wt. silver

gold = 95.23

Solution = \_\_\_\_\_

silver = N/A

Ore = \_\_\_\_\_

1195/29.166 = 40.97 A.T.  
2.578/40.97 = 0.063 oz/ton

Lime 1b/Ton  
Solution = \_\_\_\_\_

1195 grams milled 4 hours  
8 hours agitated w/240 gr.Hg.

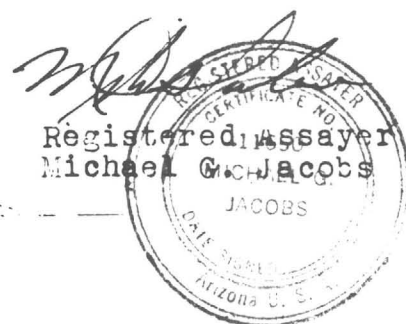
Ore = \_\_\_\_\_

Percent recovery

gold = \_\_\_\_\_

silver = \_\_\_\_\_

pH \_\_\_\_\_



DMEA LTD.

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Tucson, Arizona 85713  
602-622-0813

PRELIMINARY TESTS

SAMPLE NO. 15/1/1 conc. as received.  
Weight as received 2480 grams XXX; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.007 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

TAILS

GOLD oz/T XXXXXXX

0.005

SILVER oz/T XXXXXXX

<.05

TOTAL mg Gold 0.179

0.425

TOTAL mg  
silver N/A

N/A

Percent recovery

gold = 42.11

silver = N/A

$2480/29.166 = 85.03$  A.T.

$0.604/85.03 = 0.007$  oz/ton

2480 grams milled 4 hours

8 hours agitated w/500 grams Hg.

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

Free sodium cyanide  
lb/Ton

Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Lime lb/Ton

Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Percent recovery

gold = \_\_\_\_\_

silver = \_\_\_\_\_

pH \_\_\_\_\_

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

Iron  
mg

METALLICS

PULP

Wt. g \_\_\_\_\_

Oz/T gold \_\_\_\_\_

Oz/T

silver \_\_\_\_\_

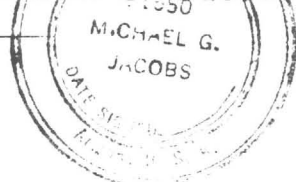
Wt. Gold \_\_\_\_\_

mg \_\_\_\_\_

Wt. silver \_\_\_\_\_

mg \_\_\_\_\_

Registered Assayer  
Michael G. Jacobs



JACOBS ASSAY OFFICE  
1435 So. 10<sup>th</sup> Avenue  
Tucson, Arizona 85713  
602-622-0813

DMEA LTD.

PRELIMINARY TESTS

SAMPLE NO. 15/1/2 conc. as received.  
Weight as received 1350 grams ~~165~~ <sup>xxx</sup>; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.310 (calc) oz/T, Silver N/A oz/T  
GOLD/200 grams xxx mg; SILVER/200 grams xxx

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

TAILS

SOLUTION

TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

Iron  
mg

METALLICS

PULP

GOLD oz/T XXXXXXX 0.002  
SILVER oz/T XXXXXXX <.05  
TOTAL mg Gold 14.300 0.092  
TOTAL mg  
silver N/A N/A

Percent recovery  
gold = 99.36  
silver = N/A

1350/29.166 = 46.29 A.T.  
14.392/46.29 = 0.310  
1350 grams milled 4 hours  
8 hours agitated w/270 gr.Hg.

Free sodium cyanide  
1b/Ton  
Solution = \_\_\_\_\_  
Ore = \_\_\_\_\_

Lime 1b/Ton  
Solution = \_\_\_\_\_  
Ore = \_\_\_\_\_

Percent recovery

gold = \_\_\_\_\_  
silver = \_\_\_\_\_

PH \_\_\_\_\_

Wt. g \_\_\_\_\_  
Oz/T gold \_\_\_\_\_  
Oz/T silver \_\_\_\_\_  
Wt. Gold \_\_\_\_\_ Wt. Gold \_\_\_\_\_  
mg \_\_\_\_\_ mg \_\_\_\_\_  
Wt. silver \_\_\_\_\_ Wt. silver \_\_\_\_\_  
mg \_\_\_\_\_ mg \_\_\_\_\_

*Michael G. Jacobs*  
Registered Assayer  
Michael G. Jacobs  
MICHAEL G.  
JACOBS  
DATE SIGNED  
Arizona U. S. A.

DMEA LTD.

JACOBS ASSAY OFFICE  
1435 So. 10<sup>th</sup> Avenue  
Tucson, Arizona 85713  
602-622-0813

### PRELIMINARY TESTS

SAMPLE NO. 16/1/1 conc. as received.  
Weight as received 247 grams ~~XXX~~; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 0.226 (cal) oz/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

#### AMALGAMATION

200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

#### CYANIDATION

200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

TAILS

SOLUTION

TAILS

#### METALLICS

200 grams - 100 Mesh

10 minute screening  
time

Iron  
mg

METALLICS

PULP

GOLD oz/T XXXXXXX 0.014

SILVER oz/T XXXXXXX 0.05

TOTAL mg Gold 1.800 0.119

TOTAL mg  
silver N/A N/A

Percent recovery

gold = 93.80

silver = N/A

$247/29.166 = 8.47$  A.T.

$1.919/8.47 = 0.226$  ox/ton

247 grams milled 4 hours

8 hours agitated w/50 gr.Hg.

Free sodium cyanide  
lb/Ton  
Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Lime lb/Ton  
Solution = \_\_\_\_\_

Ore = \_\_\_\_\_

Percent recovery

gold = \_\_\_\_\_

silver = \_\_\_\_\_

pH \_\_\_\_\_

Wt. g \_\_\_\_\_

Oz/T gold \_\_\_\_\_

Oz/T

silver \_\_\_\_\_

Wt. Gold Wt. Gold

mg mg

Wt. silver Wt. silver

mg mg

Registered Assayer  
Michael G. Jacobs



JACOBS ASSAY OFFICE  
1435 So. 10<sup>th</sup> Avenue  
Tucson, Arizona 85713  
602-622-0813

DMEA LTD.

PRELIMINARY TESTS

SAMPLE NO. 16/1/2 conc. as received.  
Weight as received 204 grams ~~XXX~~; Size, all minus 3/8 inches  
HEAD ASSAY: Gold 11.64 (caloz)/T, Silver N/A oz/T  
GOLD/200 grams XXX mg; SILVER/200 grams XXX

AMALGAMATION  
200 grams - 100 Mesh

800 cc water  
40 grams mercury  
1 gram sodium hydroxide  
Agitate 3 hours

AMALGAM

TAILS

CYANIDATION  
200 grams - 100 Mesh

800 cc water  
2 grams sodium  
cyanide  
1,2 grams lime  
Agitate 24 hours

SOLUTION

TAILS

METALLICS  
200 grams - 100 Mesh

10 minute screening  
time

Iron

mg

METALLICS

PULP

GOLD oz/T XXXXXXX

1.028

Wt. g           

SILVER oz/T XXXXXXX

0.10

Oz/T gold           

TOTAL mg Gold 74.160

7.185

Oz/T

silver           

TOTAL mg

silver N/A

N/A

Wt. Gold

Wt. Gold

mg           

mg           

Percent recovery

Free sodium cyanide  
lb/Ton

Wt. silver

Wt. silver

gold = 91.17

Solution =           

mg           

mg           

silver = N/A

Ore =           

$204/29.166 = 6.99$  A.T.

$81.345/6.99 = 11.64$  oz/ton

Lime lb/Ton

Solution =           

204 grams milled 4 hours

8 hours agitated w/40 gr.Hg.

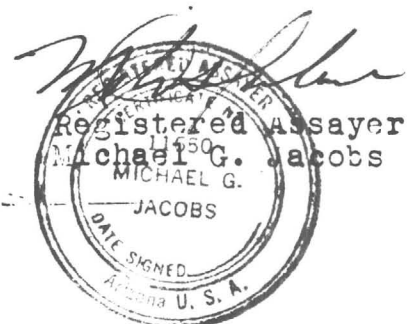
Ore =           

Percent recovery

gold =           

silver =           

pH           



1435 S. 10th AVE.

# Jacobs Assay Office

Registered Assayers



PHONE 622-0813

Certificate No.

64278

TUCSON, ARIZONA 85713

12/28 1984

Sample Submitted By Mr.

PRUDDEN

SAMPLE MARKED	GOLD Ozs. per ton ore	<del>GOLD Value per ton ore</del>	SILVER Ozs. per ton ore	<del>LEAD Per cent Wet Assay</del>	<del>COPPER Per cent Wet Assay</del>	<del>Per cent Wet Assay</del>	<del>Per cent Wet Assay</del>
TAILS 3/2/1 -10M	0.001 \$		< 0.05				
" " +10M	0.005		< 0.05				
" 4/2/1 -10M	0.012		< 0.05				
" " +10M	TRACE		< 0.05				
" 11/1/1 -10M	0.002		< 0.05				
" " +10M	0.009		< 0.05				
" 11/1/2 -10M	0.001		< 0.05				
" " +10M	0.002		< 0.05				
" 11/1/3 -10M	0.001		< 0.05				
" " +10M	TRACE		< 0.05				
" 11/1/4 -10M	0.004		< 0.05				
" " +10M	0.001		< 0.05				
" 12/1/1 -10M	TRACE		< 0.05				
" " +10M	TRACE		< 0.05				
" 12/1/2 -10M	0.002		< 0.05				
" " +10M	0.001		< 0.05				
" 12/1/3 -10M	0.001		< 0.05				
" " +10M	0.001		< 0.05				
" 13/1/1 -10M	0.003		< 0.05				
" " +10M	0.002		< 0.05				
" 14/1/1 -10M	0.001		< 0.05				
" " +10M	0.004		< 0.05				
" 15/1/1 -10M	TRACE		< 0.05				
" " +10M	0.001		< 0.05				
" 15/1/2 -10M	0.001		< 0.05				
" " +10M	TRACE		< 0.05				
" 16/1/1 -10M	0.001		< 0.05				
" 16/1/1 +10M	0.001		< 0.05				
" 16/1/2 -10M	TRACE		< 0.05				
" " +10M	0.002		< 0.05				

FIRE ASSAY

2 ASSAY FOR GASAMPLE

WET SAMPLES

Charges \$

315<sup>00</sup>

Very respectfully,



1435 S. 10th AVE.

# Jacobs Assay Office

Registered Assayers

PHONE 622-0813

Certificate No.

64258

TUCSON, ARIZONA 85713

12/27 1984

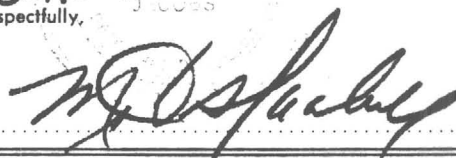
Sample Submitted By Mr.

PRUDDEN

SAMPLE MARKED	GOLD OZS PER TON	<del>X</del>	SILVER OZS PER TON	SAMPLE MARKED	GOLD OZS PER TON	<del>X</del>	SILVER OZS PER TON
① TANS 1/1/0 -10M	0.001		< 0.05	TANS 8/1/1 -10M	0.002		< 0.05
① " 1/1/0 +10M	0.009		< 0.05	" 8/1/1 +10M	0.002		0.05
② " 1/4/6 -10M	0.002		< 0.05	" 8/1/2 -10M	0.002		0.05
② TANS 1/4/6 +10M	0.002		< 0.05	" 8/1/2 +10M	0.003		0.05
③ " 1/1/10 -10M	0.001		< 0.05	" 8/1/3 -10M	0.004		< 0.05
③ " 1/1/10 +10M	0.001		0.05	" 8/1/3 +10M	0.001		< 0.05
④ " 1/2/2 -10M	0.001		< 0.05	" 9/1/1 -10M	0.001		0.05
④ " 1/2/2 +10M	0.002		0.05	" 9/1/1 +10M	0.001		0.05
⑤ " 1/2/4 -10M	0.002		< 0.05	" 10/1/1 -10M	0.018		0.05
⑤ " 1/2/4 +10M	0.003		< 0.05	" 10/1/1 +10M	0.002		0.05
⑥ " 2/1/1 -10M	0.002		0.05				
⑥ " 2/1/1 +10M	0.003		0.05				
⑦ " 2/1/2 -10M	0.002		0.05				
⑦ " 2/1/2 +10M	0.001		< 0.05				
⑧ " 3/1/1 -10M	TRACE		0.05				
⑧ TANS 3/1/1 +10M	TRACE		< 0.05				
⑨ " 3/1/2 -10M	0.002		0.05				
⑨ " 3/1/2 +10M	0.002		0.05				
⑩ " 4/1/1 -10M	TRACE		0.05				
⑩ TANS 4/1/1 +10M	TRACE		0.05				
⑪ " 4/1/2 -10M	0.001		< 0.05				
⑪ " 4/1/2 +10M	0.002		< 0.05				
⑫ " 5/1/1 -10M	0.001		< 0.05				
⑫ " 5/1/1 +10M	TRACE		< 0.05				
⑬ " 5/1/2 -10M	TRACE		< 0.05				
⑬ " 5/1/2 +10M	TRACE		< 0.05				
⑭ " 5/1/3 -10M	TRACE		< 0.05				
⑭ " 5/1/3 +10M	0.001		0.05				
⑮ " 6/1/1 -10M	0.001		0.05				
⑮ " 6/1/1 +10M	0.001		0.05				
⑯ " 6/1/2 -10M	TRACE		< 0.05				
⑯ " 6/1/2 +10M	TRACE		< 0.05				
⑰ " 7/1/1 -10M	TRACE		< 0.05				
⑰ " 7/1/1 +10M	0.002		0.05				
⑱ " 7/1/2 -10M	0.002		0.05				
⑱ " 7/1/2 +10M	0.005		< 0.05				
⑲ " 7/1/3 -10M	0.002		< 0.05				
⑲ " 7/1/3 +10M	0.002		0.05				

FIRE ASSAY  
2 ASSAY TONS SAMPLE  
SCREENING

Very respectfully,



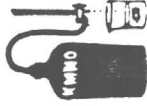
Charges \$

504.00



1435 SOUTH 10TH AVENUE  
TUCSON ARIZONA 85713

PHONE 622-0813



# Jarob's Assay Office Registered Assayers

Cert. # 64309

17 JAN 1935

Tucson, Arizona

DMEA LTD. c/o JAMES PRUDON

Sample Submitted by Mr.

Sample Marked	GOLD Ozs. per ton ore	GOLD Value per ton ore	SILVER Ozs. per ton ore	COPPER Per cent Wet Assay	LEAD Per cent Wet Assay	Per cent Wet Assay	Per cent Wet Assay	Per cent Wet Assay
"Grainy Tails"	0.016	—	2.50	<del>X</del>	<del>X</del>	<del>X</del>	<del>X</del>	<del>X</del>
"Malgamation"								
TAILS	0.071	—	5.15					
Special Bulk sample - Spentine wt as rec - 5.675 grams								
Extra sample caused to be mesh - Gravity concentrated								
Conc wt - 6 grams - Ball milled 4 hours, agitated								
8 hours with 5 grams Hg.								
Total of Gold recovered = nil								

Charges \$ 90.00

Very respectfully,

1435 SOUTH 10TH AVENUE  
TUCSON ARIZONA 85713

# Jacobs Assay Office

## Registered Assayers



PHONE 622-0813

64256

Tucson, Arizona, \_\_\_\_\_

12/20/84

Sample Submitted by Mr. \_\_\_\_\_

JAMES M. PRUDDEN

Sample Marked	GOLD Ozs. per ton ore	<del>GOLD Value per ton ore</del>	<del>SILVER Ozs. per ton ore</del>	<del>COPPER Per Cent Wet Assay</del>	<del>LEAD Per Cent Wet Assay</del>	<del>Per Cent Wet Assay</del>	<del>Per Cent Wet Assay</del>	<del>Per Cent Wet Assay</del>
H-210-5	0.017 mg Au							
H-21								
5-105-345	0.058 mg Au							
5-105-400								
H-21								
0-5-5-95	0.081 mg Au							
5-105-410								
H-34								
10-15	0.026 mg Au							
ENTIRE SAMPLE FINE BEAMED								

Charges \$

39.00

Very respectfully,

1435 SOUTH 10TH AVENUE  
TUCSON ARIZONA 85713

# Jacobs Assay Office

## Registered Assayers



PHONE 622-0813

64279  
Sample Submitted by Mr.

PRUDDEN

Tucson, Arizona.

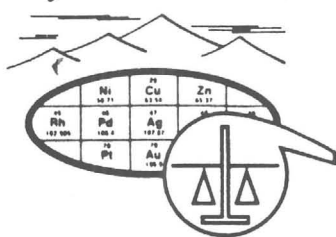
1/4, 1955  
CONC.

Sample Marked	GOLD Ozs. per ton ore	GOLD Valuation per ton ore	SILVER Ozs. per ton ore	COPPER Per cent Wet Assay	LEAD Per cent Wet Assay	Per Cent Wet Assay	Per Cent Wet Assay
H-19	0.012mg		0.30mg				
0-5	0.012mg		0.30mg				
5-10	0.012mg		0.30mg				
10-15	0.012mg		0.30mg				
15-20	0.012mg		0.30mg				
H-31	0.012mg		0.30mg				
10-15	0.012mg		0.30mg				
15-20	0.012mg		0.30mg				
H-34	0.012mg		0.30mg				
15-20	0.012mg		0.30mg				
Entire sample fine assayed							

Charges \$ 70.00

Very respectfully,

APPENDIX IV



# SKYLINE LABS, INC.

1775 W. Sahuaro Dr. • P.O. Box 50106

Tucson, Arizona 85703

(602) 622-4836

## REPORT OF SPECTROGRAPHIC ANALYSIS

JOB NO. UQX 023

January 10, 1985

BLACK SAND CONCENTRATE

A.F. BUDGE (MINING) LIMITED

Attn: Ms. Carole A. O'Brien

DMEA Ltd.

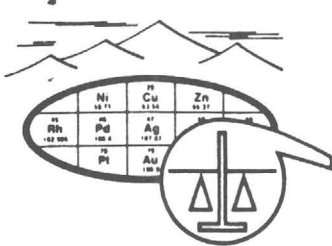
4203 North Brown Avenue, Suite F  
Scottsdale, Arizona 85251

### Analysis of 1 Black Sand Concentrate

The attached pages comprise this report of analysis. Values are reported in parts per million (ppm), except where otherwise noted, to the nearest number in the series 1, 1.5, 2, 3, 5, 7, 10, etc. within each order of magnitude. These numbers represent the approximate boundaries and midpoints of arbitrary ranges of concentration differing by the reciprocal of the cube root of ten. The 'accepted' value is considered to be within + or - 1 step of the range reported at the 68 % confidence level and within + or - 2 steps at the 95 % confidence level.

-----  
William L. Lehmbeck  
Manager

cc: Jim Prudden ✓  
Frank Millsaps



# SKYLINE LABS, INC.

1775 W. Sahuaro Dr. • P.O. Box 50106

Tucson, Arizona 85703

(602) 622-4836

JOB NO. UQX 023

PAGE 2

ITEM NO. SAMPLE NO.  
1 = BLACK SAND CONCENTRATE

ITEM 1

## ELEMENT

Fe >20%  
Ca 1.5%  
Mg .3%

Ag 150  
As <500  
B 10  
Ba 2000

Be <2  
Bi 10  
Cd <50  
Co 100

Cr 500  
Cu 300  
Ga <10  
Ge <20

La 1000  
Mn 700  
Mo 2000  
Nb <20

Ni 200  
Pb >10000  
Sb <100  
Sc 10

Sn <10  
Sr <100  
Ti 10000  
V 500

W 50  
Y 300  
Zn 700  
Zr 5000

# **Jarobz Assay Office** **Registered Assayers**

PHONE 622-0813

1/24 85

64308

Tucson, Arizona,

PRUDEN

DMET

Sample Submitted by Mr.

Sample Marked	<del>GOLD</del> Ozs. per ton ore	<del>GOLD</del> Value per ton ore	<del>SILVER</del> Ozs. per ton ore	<del>COPPER</del> Per cent Wet Assay	LEAD Per cent Wet Assay	<del>Per Cent Wet Assay</del>	<del>Per Cent Wet Assay</del>	<del>Per Cent Wet Assay</del>
3/1/1000					0.55			
4/1/2000					0.08			
7/1/2 Cent.					0.06			

7.50

Charges \$

Very respectfully,

Respectfully,  




Notes - BFD October, 1984

Abstract of NEI report on Vulture Prospect by Michael Donnelly  
March 16, 1981

#### Recommendations

High grade mineralization restricted to zones of quartz veining and adjacent wall rock. Gold in metasediments has direct correlation with intensity and extent of hydrothermal alteration; altered portions of quartz monzonite intrusive are weakly mineralized.

Best potential for gold mineralization on meaningful scale (NEI) is restricted to the intrusive body; veins near exhaustion and gold in metasediments very low grade.

IP survey revealed nothing much of value; one weak anomaly - maybe one drill hole on it.

Property of some interest, but high cost deal precludes work at this time. (Beal mentioned unfavorably.)

#### Introduction - History

Nothing much new. Two north dipping, east raking ore shoots mined; west shoot to 600 feet; east shoot to approx. 1000 feet; one stope on west shoot, 100-ft. level was approx. 80 feet wide. Noranda had 6-month walk-on at \$1,000/month. In 1970, conducted a bulk sampling program. No results given.

Present work consisted of rock chip and channel samples; detailed mapping in pits; and 20 shallow rotary holes (24-36 feet deep).

#### Regional Setting

Nothing new - Rehrig

#### Geology - Mine Area

##### Protovolcanics & sediments

Series of metavolcanics, amphibolites, fine- to coarse-grained epiclastics, quartz-pebble conglomerate and some volcanoclastics; a small lens (?) of ferruginous chert located north of pits is only chemical sediment seen. Upper greenschist metamorphism (staurolite to chlorite); some folding and pegmatites (aplite?).

Metasediments = wackes, quartz-wackes, siltites and quartzites; quartz pebble lens (?) exposed in pits; 75 feet thick in east pit; matrix strongly foliated quartz-sericite.

No bed top indicators: rock probably equivalent to Spud Mountain, or, Iron King volcanics: approx. 1760 m.y. age

#### Quartz Monzonite Porphyry Intrusive

Least understood rock in the mine area; poorly exposed; multiple facies and alteration overprint make it difficult to map.

Field evidence indicates Qm brought hydrothermal alteration to metasediments along with aplite and pegmatites; latter two very subordinate to Qm porphyry. Outcrop extent of Qmp is poorly defined; that exposed west of mine area dips under gravels to south, east and west. Mapping does show intrusive persists for 1 mile to north and northwest.

Small exposure of Qmp in pits in central part of mine area represents thin knobby apophysis of main intrusive mass. Qmp not seen underground.

#### Tertiary

Mid-Tertiary volcanics mostly andesite lavas and tuffs with some intercalated epiclastic and volcanoclastic materials. A 1929 Mining Journal article suggests 500 to 600 feet of volcanics on east side of Schoolhouse Fault in drill holes. The only Tertiary unit in the mine area is porphyritic rhyolite dike exposed in southwestern most pit.

#### Quartz veins - zones

Two large and several smaller veins provided most of historic production; sub-parallel, east-west, semi-conformable to enclosing metasediments. Two large veins tentatively traced through pit areas for 900 feet.

Veins mostly quartz-carbonate with a little pyrite, chalcopyrite and galena; native gold in flakes and leaves in association with basemetal rich portions.

All mineralized vein samples show multiple stages of brecciation, silicification and annealing in thin section and hand specimen. Silicified wall rock fragments occur in all veins and show ghost outlines; changes color of veins to blue-grey; may be a quartz vein breccia.

Absence of cross-cutting relationships prevents establishing relative ages among veins. Two barren vein events in mine area; 2-30 mm thick veins in metasediments.

#### Structure

Fabric dominated by west trending, north dipping metasediments; both Qmp intrusive and quartz veins are partly controlled by rock structural grain; 35 to 55 degree north dips.

Limited folding, northeast plunging isoclinal; wave length less than 1 meter in east wall of west pit.

### Faulting

Property in broad zone of intense north-northwest to northwest trending normal faults - shaped Vulture range.

Three episodes can be seen

1) In mine area: east-west to west-northwest trending high angle normal faults, predating quartz veins. These provided avenue for hydrothermal alteration best seen in east pit.

2) Post mineral faulting a) low angle normal faults displace quartz veins in stair step fashion; b) north-northwest trending high angle faults with right lateral movement; age relationship between (a) and (b) not known.

3) North trending Basin and Range faulting; youngest episode in area. The Schoolhouse Fault exemplifies; may have 500 to 600 feet of vertical displacement.

### Hydrothermal Alteration

All metasediments and Qpi rocks in mine area have been hydrothermally altered. Quartz-sericite-pyrite is characteristic, and is most advanced in the Qpi apophysis and in fine-grained clastic units.

As a rule, metasediment alteration is strongly developed adjacent to mineralized quartz veins; alteration decreases laterally for 10's of feet before it becomes indiscernable with greenschist metamorphism. However, this relationship breaks down when alteration is structurally controlled and alteration overlaps. Structurally controlled alteration best seen in southwestern corner of east pit.

Advanced alteration in metasediments frequently obliterates all relict sedimentary features. Fine-grained clastics are most susceptible and are usually replaced by an auriferous qtz-ser-py association. Pyrite may form 1-5% volume of rock.

### Mineralization

Rock chip geochem in conjunction with mapping shows that gold mineralization occurs in 2 distinct modes:

1) mineralized quartz veins and in their hydrothermally altered wall rocks;

2) disseminated gold mineralization in Qpi. Although copper, lead, zinc and silver are present in anomalous amounts, gold is the principal metal. Gold-silver ratios vary from 1:1 in weakly altered metasediments to 1:3.8 in strongly mineralized

quartz veins.

Base metal concentrations are low in all units, with the exception of high grade samples. Arsenic content is quite low, 15-20 ppm, and shows no relationship to gold. Gold vs. lithology plot clearly shows gold mineralization not confined to a single stratigraphic unit or rock type. Gold does vary with the intensity of alteration.

Gold mineralization in Qpi is poorly understood. Rock chip analyses suggest that Qpi in the immediate mine area is weakly gold bearing and there is correlation between intensity of alteration and gold content. Pyritic samples usually the most strongly mineralized. Gold values in Qpi range from a low of 0.04 ppm to a high of 2.2 ppm.

#### IP Survey

Designed to detect possible sulfide-rich portions of intrusive body and to delineate strongly altered metasediments (Mining Geophys. Surveys of Tucson); 4 north-south trending lines, 500-600 feet apart.

Result: weak amplitude IP response with associated weak resistivity signal on 2 adjacent lines. Anomaly described as an east-west trending dike-like response at, or near the surface that is associated with a relatively high resistivity rock of 500 ohmmeters vs. background of about 150. The significance of this resistivity contrast is not clear. Geophys. estimated that under ideal conditions the 20 ms anomaly may equate to 1% volume sulfides.

The centre of the anomaly lies between the west open pits and the central mine area and continues to the east where it is very weak on line 4, which is centered over old mine workings in central pit of mine area.

Cross sections B-B' and C-C' show the north dipping projection of the pyritic altered rocks and is a likely explanation for the IP response. Rocks (altered metasediments and Qpi) in this area contain 0.5 to 1.0% volume pyrite.

#### Discussion

This work has added a certain amount of understanding to the gold mineralization at Vulture:

- 1) altered metasediments in mine area are ubiquitously mineralized;
- 2) gold content of more than 1.7 ppm (0.05 oz/t) in metasediments generally restricted to zones of more intense alteration adjacent to quartz veins;
- 3) gold content fades laterally for 10's of feet before



ED

# The United States of America

Certificate  
No 171-

To all to whom these presents shall come, Greeting:  
Whereas Thomas H. Greenhaw Probate Judge  
of Maricopa County, Arizona Territory, in trust  
for the several use and benefit of the occu-  
pants of the Town site of Vulture City, in  
said County and Territory, according to their  
respective interests has deposited in the  
General Land Office of the United States a  
Certificate of the Register at Prescott Arizona  
Territory whereby it appears that full pay-  
ment has been made by the said Thomas H.  
Greenhaw Probate Judge as aforesaid in trust  
as aforesaid, according to the provisions of the  
Act of Congress of the 24th of April 1820 enti-  
tled "An Act making further provision for  
the sale of the Public Lands" and the Acts  
supplemental thereto including Section 2387  
of the Revised Statutes for the following  
described tract of land to wit:

To establish the N.W. corner of Vulture  
City Townsite I set a sawed pine post 6 x  
6 in. sq. & 5 1/2 feet long, marked V.C.T.S.  
N.W. cor. 20 ft. in ground, having first leveled  
the top of post and planted underneath it  
2 quart of coarse charcoal.

From this corner a witness monument  
is set high by 3 feet in diameter at base  
with 4 iron posts 4 inches marked witness  
V.C.T.S. planted firmly in centre bearing N. 81°  
03' E. variations in 1568 2607 chains and stands  
on the Summit. Hunter Peak a prominent  
conical hill having a rocky crest and ris-  
ing about 150 feet above the plain.

From this N.W. corner I run N 10° 40'  
E. Variation in 1568 10° E by observation in Polar  
with Reichen Transit



Chains 13.80 Centre of wash 25 feet wide 4 feet deep bears S. 30° 48' E.

40.00 Set a sawed pine post 6x6 in. square 5 1/2 feet long marked V.C.S. & N. cor. two feet in ground having first beveled the top of post and planted underneath it two quarts of charcoal for the S.W. corner of Vulture City Townsite.

From which a pole runs  
Tree 6 in diam. at base blazed  
and marked N. cor. V.C.S. & N. bears  
N. 44° 30' E. 113 lbs.

From this corner down  
S 79° 20' E.

Variation 14° 15' E.

Chains 10.20 Centre of wash 25 feet wide 5 feet deep bears south.

" 25.00 Top of Northern Slope of small hill.

" 40.00 Set a sawed pine post 6x6 inches square 5 1/2 feet long marked V.C.S. & S. cor. 2 1/2 feet in ground having first beveled top of post and planted underneath it 2 quarts of charred charcoal for the south-east quarter of Vulture City Townsite.

From this corner down  
N 10° 40' E.

Variation 14° 15' E.

Chains 6.00 Centre of wash 10 ft deep 30 ft wide bears S. 11° E.

" 11.55 Centre of wash 7 feet deep 25 feet wide bears S. 27° E.

" 40.00 Set a sawed pine post 6x6 in. square 5 1/2 ft. long marked V.C.S. & N. corner 5 1/2 feet in ground having first beveled top of post and planted underneath it 2 quarts of charred charcoal.



for the N.E. cor. of Vulture City Townsite.  
From this corner I run N.  $79^{\circ} 26'$  E.  
Variation  $11^{\circ} 15'$  E.

chains 7.75 Top of ridge between two small hills

" 10.25 Centre of wash 30 feet wide 11 feet  
deep bears S.  $61^{\circ}$  E.

" 13.00 Wash 25 feet wide 8 feet deep  
bears S.  $20^{\circ} 33'$  E.

" 20.15 Top of hill 25 feet high.

" 33.40 Centre of wash 20 feet wide 11  
ft deep bears S.  $2^{\circ} 35'$  E.

40.00 To N.W. corner post, the place of  
beginning, containing one hundred and  
sixty acres according to the Official Plat of  
the Survey of the said lands returned to  
the General Land Office by the Surveyor  
General, which said tract has been pur-  
chased by the said Thomas G. Greenham.

Private Judge as aforesaid in trust as aforesaid,  
Now know ye that the United States

of America, in consideration of the promises  
and in conformity with the several Acts of  
Congress in such case made and provided,  
Adopted and granted and by these presents  
Do give and grant unto the said Thomas G.  
Greenham Probate Judge as aforesaid, in trust  
as aforesaid and to his successors, the said tract  
above described, to have and to hold the same  
together with all the rights, privileges, immu-  
nities and appurtenances of whatsoever nature  
thereto belonging unto the said Thomas G.  
Greenham Probate Judge as aforesaid, in trust  
as aforesaid and to his successors and assigns  
in trust as aforesaid. Provided, That no title  
shall be hereby acquired to any mine of gold,  
silver, copper, or coal, or to any valid claim



~ E ~

claim or possession held under existing laws; and provided further, that the grant hereby made is held and declared to be subject to all the conditions, limitations and restrictions contained in Section two thousand three hundred and eighty six (2386) of the Revised Statutes of the United States, in far as the same are applicable thereto."

In Testimony whereof, Chester A. Arthur, President of the United States of America, has caused these letters to be made Patent, and the Seal of the General Land Office, to be hereunto affixed.

Given under my hand at the City of Washington the Twentieth day of June

in the year of Our Lord one thousand eight hundred and eighty-two and of the Independence of the United States the one hundred fortieth

L. S.

By the President Chester A. Arthur  
By Wm H. Cook Secretary  
S. W. Clark Recorder of the General Land Office

Head 0.072

Opi.

$(-\frac{1}{4})$

.072	-1 + 3/4	
.021	-3/4 + 1/2	
.034	-1/2 + 1/4	
.026	-1/4 + 6 mesh	
.057	-6 + 10 mesh	
.008	-10 mesh.	36%

Head 0.058

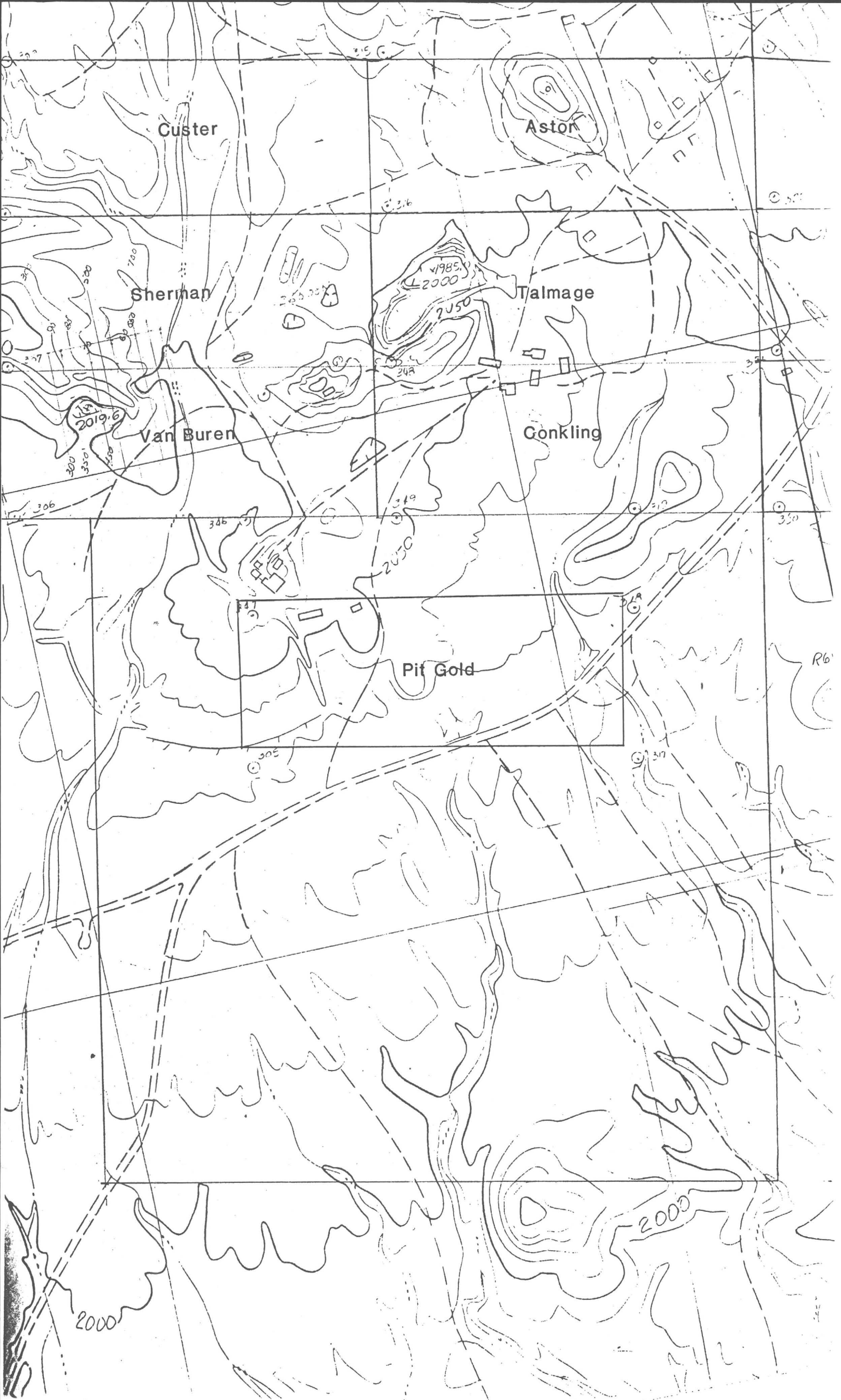
H.W.

.011	-1 + 3/4	$(-\frac{1}{2})$
.044	-3/4 + 1/2	
.014	-1/2 + 1/4	
.031	-1/4 + 6 mesh	
.015	-6 + 10 mesh	
.016	-10 mesh.	58%

Head 0.032

F.W.

.021	-1 + 3/4	$(-\frac{5}{8})$
.021	-3/4 + 1/2	
.021	-1/2 + 1/4	
.020	-1/4 + 6 mesh	
.013	-6 + 10 mesh	
.003	-10 mesh	74%



26

MS 4147

25

MS 4412

1 37.40

2511 206.442  
377

2 35.83

35

SEE SUPPLEMENTAL PLAT

36

MS 2511 206.442  
129377  
D/C

VULTURE  
CITY  
TOWNSITE

171

VULTURE  
CITY  
TOWNSITE

171

Lat. 33°48'35"N  
Long. 112°49'43"W

3 40.02 2 40.00 1 39.99 4 39.98

1919

7 29.27

5 18.46

6 10.05

2 269

1

CURRENT TO

OG

11

12

081304  
Recon NOE





2 37.51

1 44.76

MS 4147

MS 4412

MS 2511 206.442

PHX 077846  
SX Waiver  
All Min

4 12.05

5 19.26

3 18.08

36

10 30.82

129377  
D/C

MS 440 20.661 A.

MS 4409 13.517A.

MS 2522 20.661

129376  
D/C

12 29.27

11  
9.11

PHX 077846  
SX Waiver  
All Min

PHX 077846  
SX Waiver All Min

VULTURE CITY TOWNSITE

171



## Vulture Core for Metallurgical Tests

### Procedure

15 intervals (5 from each of 3 rock types: Qpi, F.W. & H.W.)  
sent to Dawson Labs, Salt Lake City

each individual interval crushed to 1"; sample cut from each  
interval for head assay; crushed  
to 20 mesh; sample cut for assay  
(200 mesh)

Composite of each rock types made of intervals selected to give  
approximate feed of 0.05 oz/t

5 kg. sample cut from each composite for bottle roll tests

Bottle roll for 72 hours to determine crushing size and  
length of leach time in column

solution checked each 24 hours  
for cyanide strength and PH

start with 10# cyanide

After bottle roll, solid-liquid separation;

liquid to assay; screen solids;  
e.g. 3/4, 1/2, 3/8, 1/4, 6 mesh,  
8 mesh, & 12 mesh; 6-7 assays

Once crushing size and leach time has been determined;

load columns (3) @ 150# each, plus 4th column of combination of  
3 composites; possible leach time: 5 days.

While column is leaching, run tests on solution again every 24 hours.

*DMEA Ltd.*  
Mineral Exploration Advice

*Ben F. Dickerson III*  
Registered & Certified Geologist  
*Carole A. O'Brien*  
Certified Geologist

7340 E. Shoeman Lane  
Suite 111 "B" (E)  
Scottsdale, AZ 85251-3335  
(602) 945-4630  
Telex: 75-1739

December 18, 1986

Mr. Harmel Dawson  
Dawson Metallurgical Labs  
5217 Major Street  
Murray, UT 84107

Re: Testing of Ore Samples  
Vulture Mine, Arizona

Dear Mr. Dawson:

I am extremely concerned by the serious time delays involved in testing of the referenced samples. Your organization was chosen to test these samples primarily because of your reputation of smoothly producing quality work.

The samples were received at your laboratory on October 28. Mr. Millsaps, our representative, advised today that assays on the bottle roll - screen tests may be available this afternoon.

This is 51 calendar days after receipt, or about 37 working days. A very large part of this time seems to have been consumed in obtaining reliable assay results.

A testing facility like yours is highly dependent on good, reliable assaying. Assayers all make mistakes. However, I have, rightly or wrongly, gained the strong impression of bumbledom at work here.

Most importantly, the amount of time consumed seems highly disproportionate to the stage of testing actually reached.

I would appreciate receiving your comments on this matter at your earliest convenience, and also your attempt to eliminate additional costly delays.

Very truly yours,

  
Ben F. Dickerson III

## Vulture Core for Metallurgical Tests

### Procedure

15 intervals (5 from each of 3 rock types: Qpi, F.W. & H.W.)  
sent to Dawson Labs, Salt Lake City

each individual interval crushed to 1"; sample cut from each  
interval for head assay; crushed  
to 20 mesh; sample cut for assay  
(200 mesh)

Composite of each rock types made of intervals selected to give  
approximate feed of 0.05 oz/t

5 kg. sample cut from each composite for bottle roll tests

Bottle roll for 72 hours to determine crushing size and  
length of leach time in column

solution checked each 24 hours  
for cyanide strength and PH

start with 10# cyanide

After bottle roll, solid-liquid separation;

liquid to assay; screen solids;  
e.g. 3/4, 1/2, 3/8, 1/4, 6 mesh,  
8 mesh, & 12 mesh; 6-7 assays

Once crushing size and leach time has been determined;

load columns (3) @ 150# each, plus 4th column of combination of  
3 composites; possible leach time: 5 days.

While column is leaching, run tests on solution again every 24 hours.

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load columns (3) @ 150# each, plus 4th column of combination of  
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# United States Department of the Interior

BUREAU OF LAND MANAGEMENT  
ARIZONA STATE OFFICE  
3707 N. 7th Street  
Phoenix, Arizona 85014

IN REPLY REFER TO:  
A MC 246423 et

(943) TR

December 5, 1985

(602) 241-5550

CERTIFIED MAIL - RETURN RECEIPT REQUESTED

## DECISION

V. M. P. Inc.  
Box 20202  
Wickenburg, AZ. 85358

Vulture #81 - A MC 246423  
Vulture #81A - A MC 246528  
Vulture #82 - A MC 246424  
Vulture #83 - A MC 246425

### Mining Claim(s) Declared Null and Void Ab Initio

Pursuant to the requirements of the Federal Land Policy and Management Act of 1976, 43 U.S.C. 1744, and the implementing regulations in 43 CFR 3833.1-2, notice(s) of location for the above-named mining claim(s) were filed for recording in the Arizona State Office of the Bureau of Land Management.

#### Date Located

September 24, 1985

#### Date Filed

December 4, 1985

The location notice(s) and accompanying map(s) show the claim(s) to be located on the following land:

Gila and Salt River Meridian, Arizona

T. 5 N., R. 6 W.  
Section 1, N $\frac{1}{2}$

T. 6 N., R. 6 W.  
Section 36, S $\frac{1}{2}$

The subject mining claim(s) are invalid and are hereby declared null and void ab initio. The Bureau of Land Management public records show the lands were not open to location of mining claims at the time of their location.

The land has been patented without a reservation of minerals to the United States. Therefore, the lands are not subject to location under the general mining laws.

"Land which has been patented without a reservation of minerals to the United States is not available for location of mining claims, and mining claims located on such land after it is so patented are null and void ab initio," Ariel C. MacDonald, et al, 52 IBLA 384 (1981).

"Mining claims located on lands which are closed to mineral entry are null and void from their inception as a matter of law, and no property rights are created thereby. Therefore, no contest proceeding notice, or hearing is required preliminary to a decision holding that such claims are invalid." John A. Ross, Maxine Lidke, 73 IBLA 16 (1983).

An appeal from this decision may be taken to the Interior Board of Land Appeals, Office of Hearings and Appeals, in accordance with the regulations in Title 43 Code of Federal Regulations (CFR), Parts 1 and 4, and the enclosed Form 1842-1. If an appeal is taken, the notice of appeal must be filed in the Arizona State Office of the Bureau of Land Management, P.O. Box 16563, Phoenix, Arizona 85011, within thirty days from the receipt of this decision. Do not send the appeal directly to the Board. The appeal and case history file will be sent to the Board from this office. Within thirty days after filing the notice of appeal, file a complete statement of the reasons why you are appealing. This must be filed with the U.S. Department of the Interior, Office of the Secretary, Board of Land Appeals, 4015 Wilson Boulevard, Arlington, Virginia 22203. If you fully stated your reasons for appealing when filing the notice of appeal, no additional statement is necessary. Additionally, within fifteen days after each document is filed, the regulations also require the appellant to serve a copy of the notice of the appeal, statement of reasons, written arguments or briefs on the Field Solicitor, U.S. Department of the Interior, 505 N. 2nd St., Suite 150, Phoenix, Arizona 85004. To avoid summary dismissal of the appeal, there must be strict compliance with the regulations.

If no appeal is taken, this decision constitutes final administrative action of this Department as affects the mining claim(s). No appeal, protest, or petition for reconsideration will be entertained from this decision after the appeal period has expired.

/s/ Thomas E. Reitmeyer

Thomas E. Reitmeyer  
Acting Chief, Branch of Lands  
and Minerals Operations

Encl: Regulations  
Appeal Procedures  
Form 1842-1

P.S. In addition, other claims i.e, B-lan 23, Vulture 32, 47 thru 52, 95 thru 97, Desert 118 thru 120, 122 thru 124 are partially located on lands in which the United States has no Federal mineral ownership. The BIM does not issue decisions on claims that are void in part. However, you have acquired no rights on these lands.

cc: Del Tierra Engineering





# United States Department of the Interior

## BUREAU OF LAND MANAGEMENT

### ARIZONA STATE OFFICE

3707 N. 7th Street  
Phoenix, Arizona 85014

IN REPLY REFER TO:  
A MC 246423 et al

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/s/ Thomas E. Reitmeyer

Thomas E. Reitmeyer  
Acting Chief, Branch of Lands  
and Minerals Operations

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cc: Del Tierra Engineering



V.M.P., Inc.  
Wickenburg, Arizona 85358

Mr. Budge

Please find enclosed, copies of my letters  
to Mr. Dickerson.

It is my sincere desire to settle this matter  
in a fair, equitable, and gentlemanly manner — however —  
Mr. Dickerson persists in making the situation totally  
one-sided — AND I might add — At both our expenses

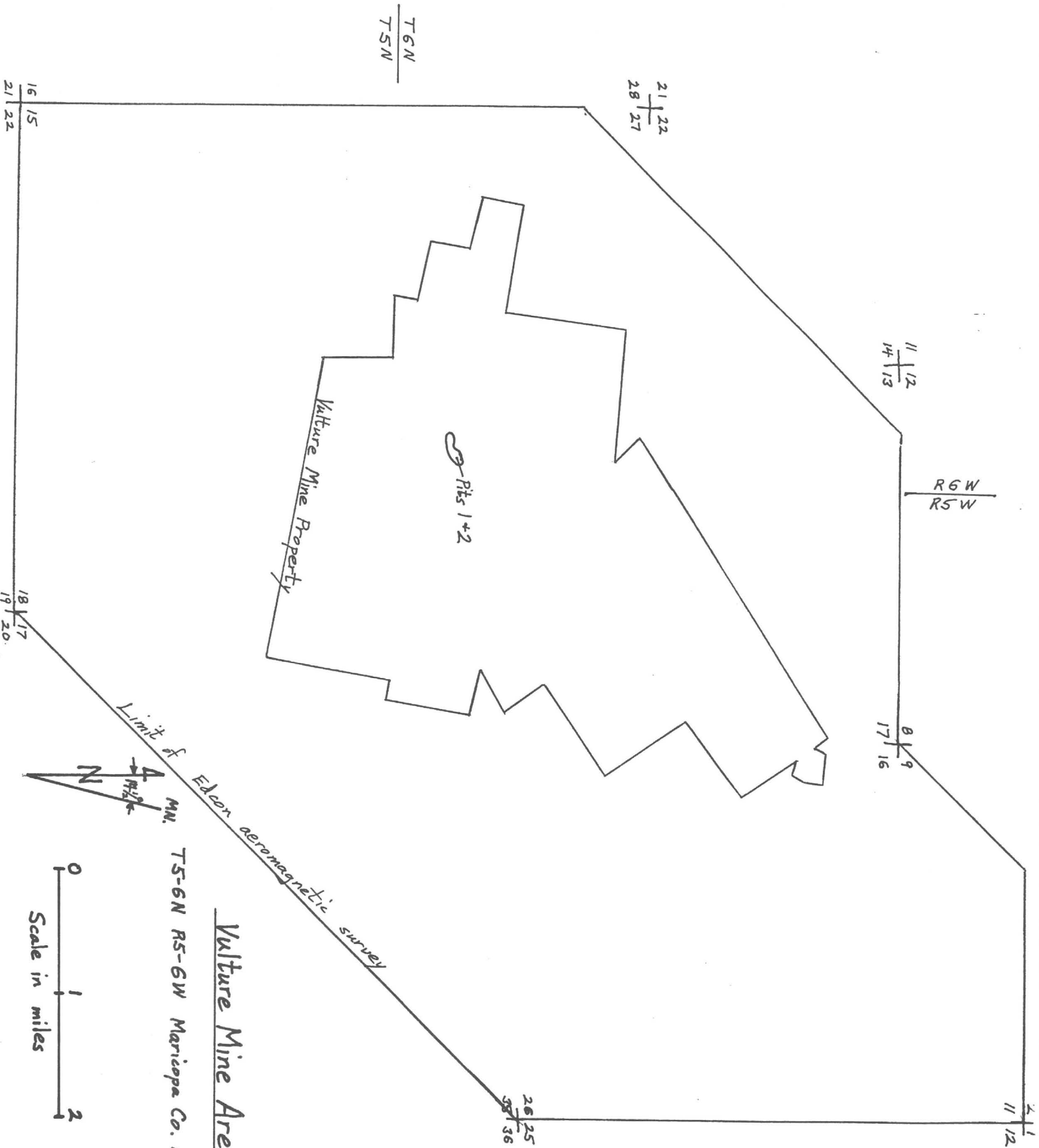
Hopefully my letters will help you to understand my  
position in this matter — My references to over charges  
ARE not unfounded.

As two businessmen we should be able to resolve  
this minor problem and proceed to the MAJOR objective  
— "PRODUCTION" —

Please feel free to contact me next time  
you are in the states —

Sincerely,  
Larry W. Budge

P.S. Have an enjoyable holiday season — AND A profitable New Year.



# Vulture Mine Area

TS-6N RS-6W Maricopa Co. Arizona

DON WHITE  
March 14, 1986

Ben & Carole,

Friday 3-14-86

This is the outline of the aeromagnetic survey area as we discussed on the phone. No time today to put the geology notes on it to explain the changes but I'll do that soon — at least so that Chuck & I can review it together when I see him Wed. afternoon in Tucson.

Don

Don White  
521 East Willis St.  
Prescott, AZ 86301  
602/778-3140

March 14, 1986

Fleetwood R. Koutz  
Geologist, Southwest Exploration Div.  
ASARCO, Inc.  
P.O. Box 5747  
Tucson, AZ 85703

Dear Fleet,

Attached is the data on U.V.X. drilling. It may be useful in your review of the data you already have. Of course that little summary report is only sketchy in that it contains  $8\frac{1}{2}$ " x 11" graphics and no drill logs. If and when you care to make a more detailed review we shall be happy to provide copies of pertinent level plans and cross sections (both 1" = 40') and drill logs (1" = 40' and/or 1" = 20') and chert detail logs (1" = 10').

Perhaps we shall be able to meet at the A.G.S. symposium later next week.

Regards,



Don White  
Geologist, C.P.G.

Enclosure

cc: Ben F. Dickerson, III ✓

DW:sk



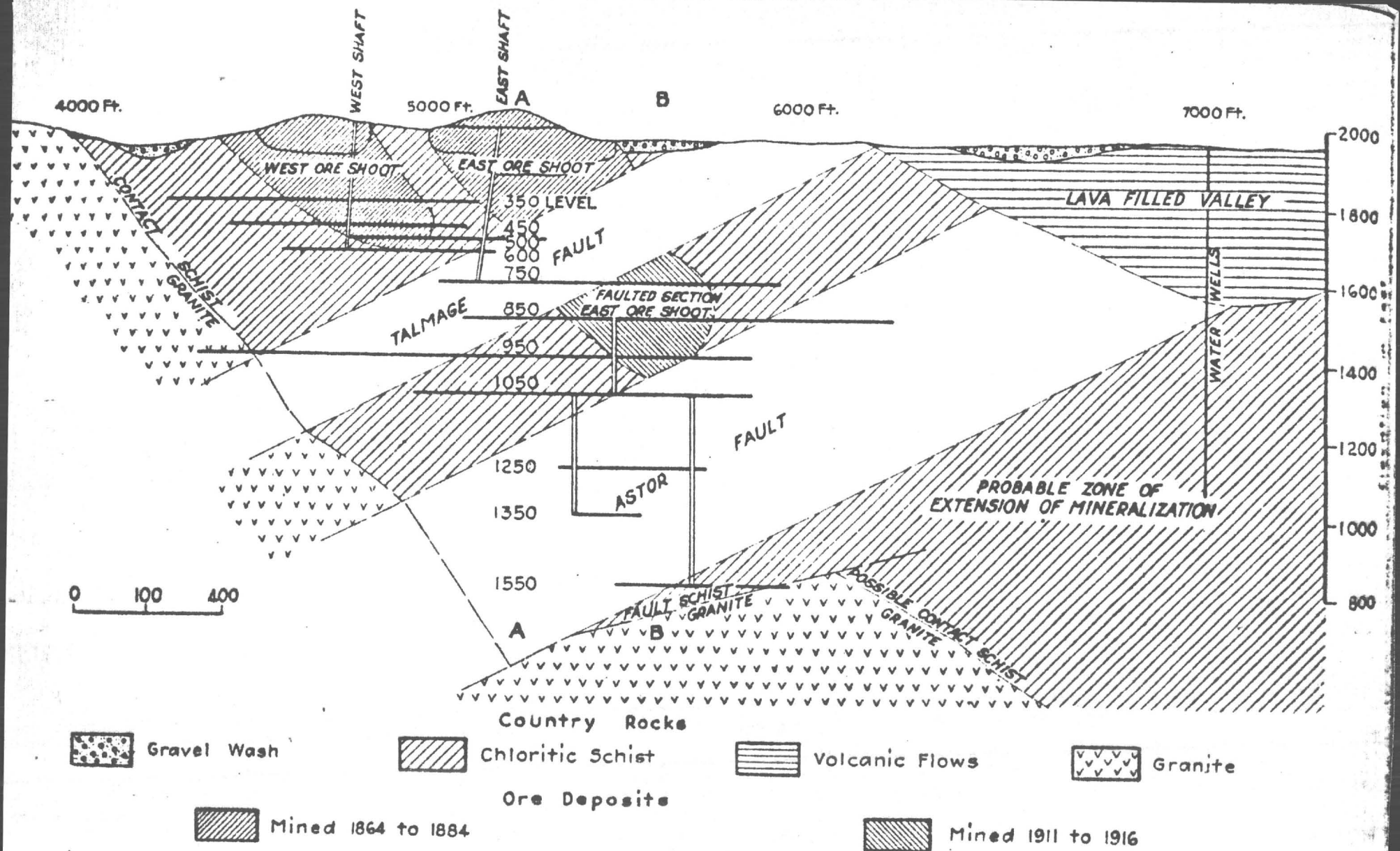


FIG. 1. LONGITUDINAL VERTICAL PROJECTION OF THE VULTURE MINE WORKINGS

Not

Scanned

(Some duplicates in adjacent  
Scanned folder)

---

Don White  
521 East Willis St.  
Prescott, AZ 86301  
602/778-3140

February 4, 1986

Charles L. Elliot  
ELLIOT GEOPHYSICAL CO., INC.  
4653 East Pima St.  
Tucson, AZ 85712

Dear Chuck,

Shipped by bus yesterday (Monday) were two boxes containing the thirty Vulture area rock samples I collected as a suite for magnetic susceptibility testing. The Greyhound bus people are supposed to phone you regarding their arrival. I'm sorry I couldn't send them U.P.S. for your convenience but our local office doesn't pick up and is only open three hours each of three days per week, never when I can get there.

Attached is a brief log of what the hand specimens look like megascopically, some interpretation of what they represent, and their grid location (the grid is the one on the 200-scale topo map Carole O'Brien sent you). Don't bother plotting their locations though, for I will get you a copy with that already done. The rock suite represents the Precambrian, Laramide (?), and Tertiary rocks of import to us.

I ran trial magnetometer lines over various exposed and covered features as a sort of orientation and test of the suitability of magnetics. I was not too enthused. The exposed faults are too narrow to be resolved, the alluvial covered faults are effectively masked, the qpi sill is not distinguished magnetically, even where outcropping, and the qpi stock to ~~pe~~ country rock contact is only vaguely suggested by the magnetics. The chance of resolving a narrow apophysis beneath basalt, ash, and granitic alluvium seems slight.

These judgements are based upon the raw data only. I am just now plotting up the profiles and will either mail them or bring them when we rendezvous.

Give me a call if anything notable comes out of the magnetic susceptibility testing. Let's touch base this weekend regarding our rendezvous at the Vulture next week. I tentatively plan to be there Monday, Feb. 10 for another matter, and as many days thereafter as necessary for the VLF work.

Regards,



Don White  
Geologist

DW:sk

Attachments

cc: Ben F. Dickerson, III and Carole A. O'Brien

VULTURE ROCK SUITE  
FOR MAGNETIC SUSCEPTIBILITY TESTING

- 1 qpi stock - Beige, white, & light gray, coarse grained granite porphyry with clear quartz porphyroblasts up to 1/4" composing about 50% of rock. Balance is feldspar, mostly orthoclase, some biotite, and muscovite and about 5% scattered iron-stained spots of an unidentified, oxidized, ferruginous mineral (pyrite-?) preferentially surrounded by a halo of white feldspar. Massive, unfoliated. Trace manganese on feldspar faces. Some carbonate (siderite-?) as disseminations around the iron stained and feldspar-rich areas.  
25,850 N - 218-770 E
- 2 qpi stock - Same as 1 except for more iron-stained spots (looking more like relict pyrite and casts thereof) up to 10%. More coarse grained, some quartz porphyroblasts approaching 1/2 inch in diameter.  
25,850 N - 218,600 E
- 3 Tertiary dike rock - Beige to light brown-gray, fine to coarse grained, massive, feldspar porphyry. White subhedral orthoclase porphyroblasts up to 1/4" (30%) and clear quartz phenocrysts up to 1/10" (20%) are set in a very fine grained groundmass. Trace casts of pyrite euhedra up to 1/10". Some carbonate which may be primary or secondary. This rock forms dikes up to 40' thick within the qpi stock. Possible trachyte composition.  
26,450 N - 218,650 E
- 4 Intrusive breccia - Black and white, coarse grained, breccia of about 50% subangular qpi fragments up to 4" ( $\bar{x} = 1/2"$ ) healed by specular hematite (forming a 1/4" thick rim to the rock fragments) and vug-filling euhedral quartz. Some casts after coarse pyrite (up to 1/4"). This rock probably doesn't occur in more than a few-foot thicknesses along faults within the qpi stock.  
26,800 N - 218,700 E
- 5 Tertiary dike rock - Same as sample 3 but slightly more pink color groundmass. Possible latite composition.  
26,750 N - 218,950 E
- 6 qpi - Same as sample 1.  
26,600 N - 219,050 E
- 7 qpi sill - Pale green variant (trace chlorite-?) of sample 1. Also contains a 1/4" thick white quartz vein. Similar to sample 1 but taken from the sill-like apophysis north of pit 3 rather than from the stock's core.  
26,400 N - 220,430 E
- 8 pC schist - Brown, green-gray, very fine grained, crudely foliated, quartz-sericite-carbonate-hematite schist with cross-cutting and conformable veinlets (up to 1/2" thick) of calcite, siderite, and quartz. Foliation probably obscured by hornfelsing as a result of qpi sill within 20 ft. north.  
26,350 N - 220,400 E

- 9 Tertiary rhyolite dike - White to very palest green very fine grained intrusive dike rock. Massive, hard (H about same as steel nail) and characteristically contains abundant manganese dendrites on virtually every fracture surface. Some hairline quartz veins cutting at various orientations. Well displayed in walls of pit 3.  
26,330 N - 220,350 E
- 10 pE schist - Gray green, very fine grained, faintly foliated, quartz-chlorite-sericite schist with trace carbonate, some conformable quartz veins. Partially hornfelsed by proximity to qpi sill. Collected from north wall of pit 3.  
26,270 N - 220-400 E
- 11 pE schist breccia - Gray green, very fine grained, foliated, quartz-sericite-chlorite schist rock fragments in a slightly more siliceous, unfoliated qpi-like matrix. Some iron staining and carbonate. From south wall pit 3.  
26,200 N - 220,450 E
- 12 pE gneiss - Black and white, banded, fine to medium grained, quartz-amphibole-sericite-biotite gneiss with some iron stain. This rock is hanging wall to the qpi north of pit 3.  
26,470 N - 220,400 E
- 13 qpi sill - Like sample 1 except for smaller, less distinct quartz phenocrysts. This is a more quickly chilled, marginal facies from the hanging wall of the sill north of pit 3.  
26,600 N - 220,380 E
- 14 Quartz - White, massive "bull" quartz thought to be vein material within and emanating from the qpi sill. Includes adhering chunks of massive sericite along what were probably walls of the vein. This sample from the gold-barren lense northwest of the mine area.  
26,760 N - 220,300 E
- 15 pE schist - Gray-green, very fine grained, foliated, quartz-chlorite-sericite-hematite-carbonate schist. Foliation preserved by being at least 200 ft hanging wall to the qpi north of pit 4.  
26,700 N - 220,600 E
- 16 qpi sill - Very pale green, white silicified, sericitized, and argillized variant of sample 1. Contains anastomosing quartz veins from hairline size to 1/2" thick, all with diffuse contacts as though the silica has penetrated the rock. This sample from cut between pit 2 and west incline.  
26,180 N - 221,340 E
- 17 pE schist - Gray-green, very fine grained, poorly foliated, silicified and hornfelsed and pyritized quartz-chlorite-sericite schist. Quartz veins up to 1/2" thick, generally conformable. Pyrite euhedra up to 1/4" both in the schist and quartz. This is the immediate footwall to the qpi sill of pit 2 (sample 16)  
26,160 N - 221,350 E

- 18 p€ schist - Iron stained silver gray, very fine grained, thick foliated, quartz-sericite-chlorite schist. This sample about 100 ft. footwall to qpi sill of pit 2 (sample 16).  
26,060 N - 221,370 E
- 19 Sulfidic qpi - Similar to sample 16 but including about 3% pyrite, 3% galena, both as euhedra up to 1/4". Some yellow sulfur stain. From west end pit 1.  
26,250 N - 221,600 E
- 20 p€ schist - Gray-brown, iron stained, very fine grained, poorly foliated, hornfelsed (?) sheared, quartz-chlorite-sericite schist. This is immediate hanging wall to qpi sill of pit 1 (sample 20).  
26,320 N - 221,600 E
- 21 Calcite - Light gray, very coarse grained, massive sparry calcite occurring as fault filling up to 2 ft. thick. This sample from Talmage fault in east wall of pit 1.  
26,500 N - 221,980 E
- 22 Fault breccia - Coarse, angular breccia fragments of same composition as sample 20 within sparry calcite (like sample 21) of fault zone. This sample from dump around shaft northwest of old schoolhouse (shaft sunk on fault which is a splay off the Schoolhouse fault).  
27,560 N - 222,360 E
- 23 p€ diabase (?) - White and light gray, medium to coarse grained, foliated plutonic textured, quartz-feldspar-sericite-biotite ± chlorite. Contact relationships not found but texturally and mineralogically this is like the qpi but foliated enough to be p€. Occurs west of the old schoolhouse.  
27,600 N - 222,270 E
- 24 p€ amphibolite - Black, fine grained, faintly banded to massive, hornblende-quartz-feldspar gneiss with pyrite occurring as iron stained clusters of euhedra up to 1/10" surrounded by a feldspar-rich halo.  
27,500 N - 222,200 E
- 25 Tertiary ash flow - Pink-gray, medium grained, massive, unfoliated, equigranular quartz-orthoclase-plagioclase ash flow rock. Collected from dump of Douglas shaft.  
27,650 N - 223,400 E
- 26 Tertiary basalt - Gray, fine to medium grained, massive and fragmental basalt or andesite flow breccia and/or ash fall. Now contains about 20% sparry calcite. Collected from dump of Douglas shaft.  
27,650 N - 223,400 E
- 27 p€ amphibolite - Dark gray, fine grained, very faintly foliated, quartz hornblende-sericite-gneiss with hematite and carbonate alteration now pervasive. This is the p€ rock at the bottom of the Douglas shaft.  
27,650 N - 223,400 E

- 28 Tertiary volcanics - White, fine to medium grained, banded, crystal lithic tuff and ash fall. Bands about 1/2" thick alternate from a crystal tuff (hard, vitrious) to a poorly sorted pumice-rich ash with abundant open space. Outcrops on twin hills in W1/2SW1/4Sec1, one mile southwest of Yulture mine.
- 29 Tertiary volcanics - Same as sample 28 but all ash and pumice. Same location as sample 28.
- 30 p6 gneiss - Light gray, fine to medium grained, faintly foliated, dense, hard, quartz-feldspar-magnetite-biotite meta-intrusive (?). Noticeably magnetic. Collected from outcrop in wash in NW1/4Sec 2 T5N R6W (AZ minerals under prospecting permit to Hunt, Ware, and Proffett).



Carde

M E M O

TO: Carole A. O'Brien, Anthony F. Budge  
FROM: Don White  
DATE: Sept. 28, 1987  
SUBJECT: Exploration ideas and opportunities near the Vulture Mine.

Compilation of the Vulture paper for the January 1988 S.M.E. meeting accompanied by careful study of Bill Rehrig's data has allowed me to recognize another Vulture-stock-like pluton worthy of exploration. Also, a property just two miles east of Caballeros Peaks (and the east edge of the Vulture claim block) has been brought to my attention with several 0.1 oz/t Au drill intercepts.

Our understanding of the genesis of the Vulture lode is summarized in the enclosed SME paper just completed. What is critical is the Vulture stock (the "quartz porphyry") and its apophysis, now dated as  $85 \pm 3$  m.y. or latest Cretaceous/earliest Laramide. The Vulture claim block encompasses perhaps one third of the Vulture stock and perimeter, mainly the northeast and east sides. The Vulture lode dominates the area of outcrops. This spring's exercise with the ground magnetics and reverse circulation drilling was an attempt to locate other qp apophyses buried by Quarternary pediments. Three such targets were drilled with all three apparently being low-mag. interbeds within the Proterozoic stratigraphy. While that is disappointing, it doesn't mean other such targets aren't Vulture-like lodes. The concept may be applied to any perimeter of any pluton of similar composition and age to that of the Vulture stock. Study of a regional map by Rehrig, et. al. 1980 (colored copy attached) highlights a total of three such plutons; the Vulture stock, the Cañon City stock, and the more northerly Hartman Wash stock (all my names). All are coarse grained quartz monzonite with quartz porphyroblasts or the so-called "qp".

The Cañon City patent is plunk in the middle of the Cañon City qp body, unlike the Vulture lode area which is off along a sill. Thus the relatively lesser size and grade of the Cañon City gold occurrence to that of the Vulture. The all-important margins of the Cañon City stock are virtually unexposed because of faulting and Tertiary volcanic and Quarternary pediment cover. I propose we devote some careful thought on how to detect qp apophyses beneath the various cover types. Geophysics (including seismics, magnetics, radar) geochemistry (incl. Tellis Gold's "sniffer") and drilling should all be considered.

The Hartman Wash qp occurrence ought to be prospected. It is in an area of no reported workings (15' Vulture topo quad). A day or two of reconnaissance traverses and sampling would tell whether it is the right sort of pluton or not.

Almost coincident with your queries regarding higher grade hard-rock gold reserves at the Vulture, I have had a submittal that could fit the bill. It is the ex-Labradex property just east of Caballeros Peaks, about a 6-mile haul from the Vulture. Labradex is no more, but spun the property off to its ex-staff. They, in turn, have retained REDCO of Reno to peddle it for them.

Carole A. O'Brien, Anthony F. Budge  
Exploration ideas and opportunities...  
September 28, 1987  
Page 2

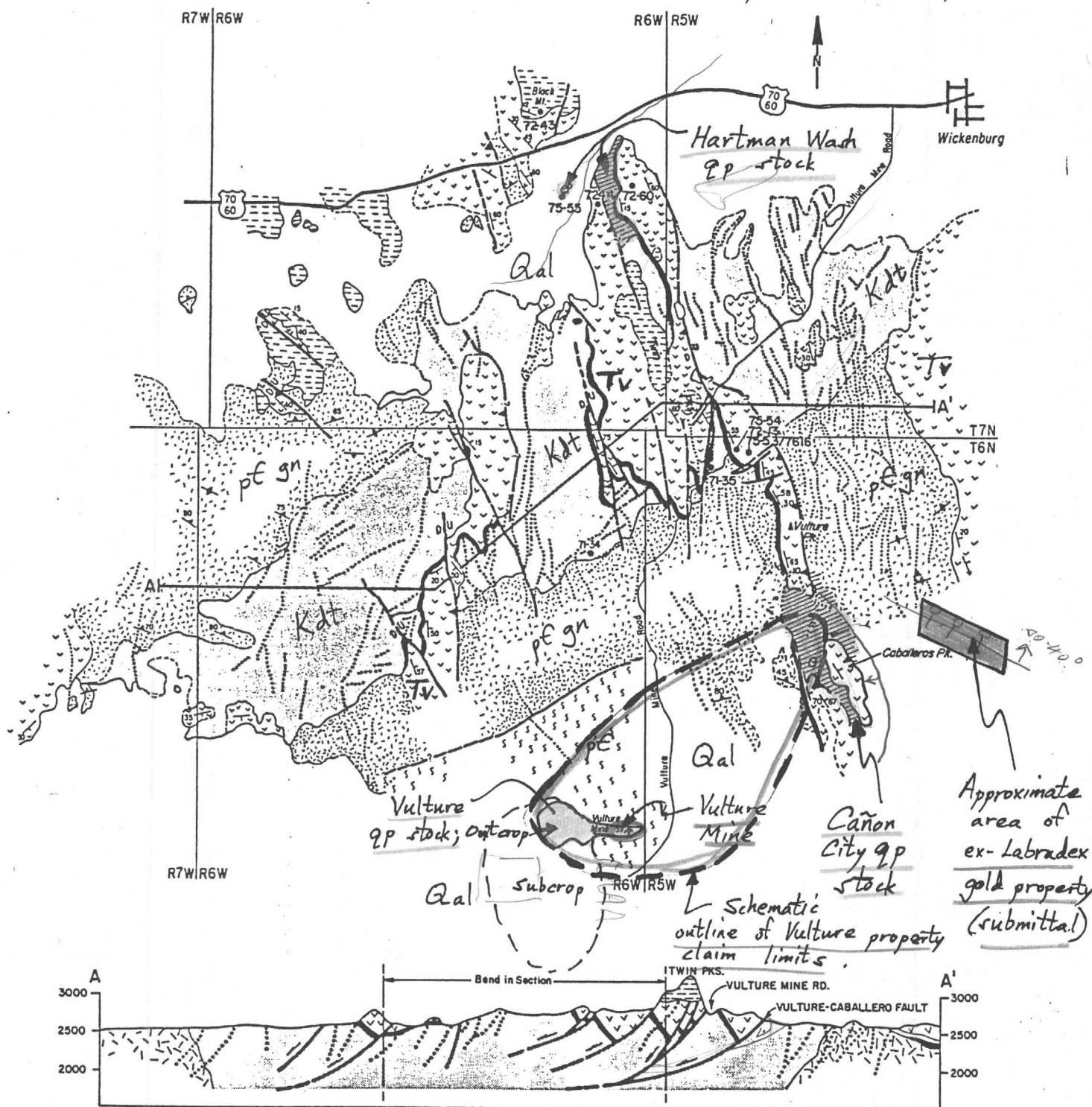
Labradex drilled a number of holes along a 3,000 foot structure, intercepting two areas of gold over 0.1 oz/t and over 10 feet thick. Stripping ratios would be similar to the Vulture but there is hope the foliated rock would crush more easily and leach with better recoveries than the Vulture q.p.

I believe the Labradex property has potential for at least two areas of 10,000 to 100,000 s.t. each, at 0.1 oz/t. It was looked at recently by Noranda, including several additional holes, but was too small and erratic for them.

The ex-Labradex owners are looking for an N.S.R. royalty. REDCO wishes to be retained for the exploration work plus a royalty on any finds, but all is negotiable. Seems to me it offers an opportunity to beef up the reserve total for any operation at the Vulture. Let me know if a field visit or study of their data is warranted.

DW:sk

Modified from:  
Rehrig, Shafiqullah, and Damon, 1980  
by Don White - Sept. 1987



Map showing three identified quartz porphyry (qp) stocks and area of ex-Labrodex property with respect to Vulture Mine and claim block.



for the N.E. cor. of Culture City Transits.

Some this corner Land N. 79° 20' E.  
Variation 11° 15' E.

chains 775 Top of ridge between two small hills

10.25 Crest of Muck 20 feet wide & 4 feet  
deep bars S 61° E

13.00 Muck 25 feet wide & 8 feet deep  
bars S 20° 33' E

20.75 Top of hill 25 feet high.

33.25 Crest of Muck 20 feet wide &  
4 feet deep bars S 2° 25' E

44.00 To N.W. corner post, the place of

beginning, containing one hundred and  
sixty acres, according to the Official Plat of  
the Survey of the said lands returned to  
the General Land Office by the Surveyor  
General, which said tract has been sur-  
veyed for the said Thomas H. Greenhaw.

Private Order as aforesaid, in trust as aforesaid,  
Now recite, That the United States

of America, in consideration of the premises  
and in conformity with the several Acts of  
Congress in such case made and provided,  
do give and grant unto the said Thomas H.  
Greenhaw Private Order as aforesaid, in trust  
as aforesaid and to his successors, the said tract  
shall be divided into two and to hold the same  
together with all the rights, privileges, immu-  
nities and appurtenances of what so ever nature  
appertains belonging unto the said Thomas H.  
Greenhaw Private Order as aforesaid, in trust  
as aforesaid and to his successors and assigns  
in trust as aforesaid. Provided, That no title  
shall be hereby acquired to any mine of gold  
silver, copper, or coal, or to any other mineral

claim or possession held under existing  
laws, and provided further, That the grant  
hereby made is held and declared to be  
subject to all the conditions, limitations  
and restrictions contained in Section two  
thousand three hundred and eighty six (2386)  
of the Revised Statutes of the United States,  
in far as the same are applicable thereto.

Do testimony whereof,  
Charles A. Arthur, President of the United  
States of America, has caused these letters  
to be made Patent and the Seal of the General  
Land Office, to be hereunto affixed.

Given under my hand at  
the City of Washington the  
twentieth day of June  
in the year of Our Lord  
one thousand eight hundred  
and eighty two and of the  
Independence of the United  
States the one hundred and  
sixth.

L. S.

By the President Charles A. Arthur  
By Wm. H. Cook Secretary  
L. M. Clark Recorder of the General Land Office





LAW OFFICES  
TWITTY, SIEVWRIGHT & MILLS  
1700 TOWNEHOUSE TOWER  
100 WEST CLARENDON  
PHOENIX, ARIZONA 85013

HOWARD A. TWITTY  
RALPH B. SIEVWRIGHT  
JOHN F. MILLS  
N. DOUGLAS GRIMWOOD  
ANDREW R. ALEX  
ROGER K. FERLAND

TELEPHONE 248-9424  
AREA CODE 602

March 11, 1982

RECEIVED MAR 15 1982

Mr. Russ Walker  
Zortman/Lancusky Mining Company  
P. O. Box 1904  
Wickenburg, AZ 85358

RE: Vulture City Townsite

Dear Mr. Walker:

I have received the townsite patent for the Vulture City Townsite issued by the United States on June 20, 1982. This copy of the patent was delivered to us by one of your representatives. In the patent, the provision is made as follows:

"Provided, that no title shall be hereby acquired to any mine of gold, silver, cinnabar, or copper or to any valid mining claim or possession held under existing law; . . ."

At the time of the issuance of this townsite patent, the matter of mineral rights in land granted by a townsite patent was unclear. It wasn't until the U.S. Supreme Court decision in 1885 was there any authoritative opinion by the U.S. Supreme Court given with respect to the rights of a townsite patentee to mineral lands. In the case of Deffebach v. Hawke, 115 U.S. 392, decided in 1885, the U.S. Supreme Court considered the effect of mining claims located on land held under the townsite laws.

Enclosed are 5 pages from the American Law of Mining discussing townsite entries. Also enclosed is a copy of an Arizona territorial decision by the Supreme Court of Arizona involving the town-site cases.

We think the effect of the Court decisions is that lands known to be valuable for minerals immediately prior to a townsite entry are excluded from the townsite entry, even though it may be included in a townsite patent. On the other hand, land included in a townsite patent that was not known to be valuable for mineral deposits are with its minerals included in the townsite patent.

Because of this uncertainty, we suggest that it would be desirable to have a title company make a title search of the history of this townsite patent. Such a search can be made by the title company more efficiently and less expensively than by an attorney or a land man. If such a search is authorized, we would employ Transamerica Title to make this search.

TWITTY, SIEVWRIGHT & MILLS

By

HAT/jel

Enclosures

cc George Hennessey w/encl.



§ 3.18 FEDERALLY RESERVED MINERALS IN FEE LANDS 510

and the state elected to select lieu land pursuant to the Act of 1891. The state had filed in 1912 a list of selected lands in full conformance with the directions of the Secretary of Interior, which was accompanied by requisite proofs and the proper fees. The section of land selected was withdrawn in 1914 as possible oil land, and the land department subsequently refused to approve the selection unless the state prove it was not known to be valuable for oil. The court specifically held that mineral character was to be tested by known conditions at the time the state had done all necessary on its part, and further indicated that school grants were to be liberally construed.

The foregoing decision reversed previous rulings by the land department that a state acquired no vested right in lieu selections until approved by the Secretary.<sup>6</sup> However, if a state has accepted a patent to lieu selections containing a reservation of minerals, the land department has taken the position that acceptance of the patent constitutes a waiver to an unrestricted patent although it subsequently develops that the department's position had been erroneous and the patentee was lawfully entitled to the unrestricted patent.<sup>7</sup>

§ 3.18 Townsite Entries. Of the many provisions, both general and special, that have been enacted for the entry of townsites upon the public lands, the first of significance to mining interests were the acts passed during the 1860's.<sup>1</sup> With regard to the character of land available for townsite entry, the Act of March 3, 1865<sup>2</sup> provided that title to town lots would be subject to recognized possession of mineral veins. Two years later legislation provided that no title to any mine of gold, silver, cinnabar, or copper, or to any valid mining claim or possession held under existing laws, could be acquired under a townsite entry.<sup>3</sup> In the leading case of *Deffebach v.*

<sup>6</sup> Robinson (1921) 48 LD 384.

<sup>7</sup> United States v. New Mexico (1933) 54 ID 175; Robinson (1921) 48 LD 384, 387.

§ 3.18 <sup>1</sup> In general see 43 USC §§ 711-731 (1958) and 43 CFR Part

255 (1958 Supp) for existing townsite laws and regulations.

<sup>2</sup> 13 Stat 530 (1865), 43 USC § 717 (1958).

<sup>3</sup> 14 Stat 541 (1867), 43 USC § 722 (1958).

*Hawke*<sup>4</sup> the Supreme Court considered the foregoing act in the light of the general mining laws of 1866 and concluded that no title could be acquired under the townsite laws to land known at the time of sale to be valuable for its minerals of gold, silver, cinnabar, or copper. It was further stated that where a mining claim had been initiated and pursued to final certificate prior to any entry of the townsite pursuant to law, there was no legislative authority for the inclusion in the mining patent subsequently issued of a reservation excluding from the patent's operation all town improvements on the surface, as the mining patent carries with it the title to the surface included within the lines of the mining location. The certificate of purchase issued the mining claimant was said to be equivalent to a patent as to third persons and that the occupation of the town prior to any formal townsite entry initiated no rights which antedated those of the mineral claimant.<sup>5</sup> At the time of formal townsite entry the land was found to be known mineral land so that no rights were acquired thereby under the townsite law. The decision reiterated the view that unoccupied land within a townsite entry was not exempt from location under the mining law.

In *Davis v. Weibbold*<sup>6</sup> the Supreme Court directly considered whether the prohibition in the townsite law against acquisition of title to any mine was a reservation of such mines to the United States. It was held that the Act merely prohibited the passage of title under the town site laws to mines which were known to exist upon the issue of the townsite patent. If not known to exist at that time, a townsite patent passes all title of the United States to the land, thereby becoming private property to which the mining laws are inapplicable, so that the title of purchasers will not be disturbed by subsequent discoveries. The Court described the basic

<sup>4</sup> (1885) 115 US 392.

<sup>5</sup> But cf. *Bonner v. Meikle* (D Nev 1897) 82 Fed 697 (the court here recognized a possessory right by a townsite unofficially founded on public land; however, the asserted rights of the mining locator appear to

have failed for the lack of a valid mineral discovery, which was treated as essential to the success of a mining claim as against occupants of a town).

<sup>6</sup> (1891) 139 US 507.

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rule when it said that where "the mining patentee's rights antedated those of the occupants under the town-site law, and wherever such is the case, his rights will be enforced against the pretensions of the townsite holder; but where the latter has acquired his rights in advance of the discovery of any mines and the initiation of proceedings for the acquisition of their title or possession, his rights will be deemed superior to those of the mining claimant."<sup>7</sup>

While the relative priority of a mining patentee or townsite patentee is an easy matter to state, it is not clear from the decisions just what factual evidence was necessary to constitute an exception of land under the townsite law. The courts have used the terms "lands known to be valuable for minerals," or "for mineral deposits," and "known mines," or "land containing known mines," as equivalent in meaning under the townsite laws.<sup>8</sup> In one case it was stated that a town took no title when the evidence shows that the existence of a mineral-bearing vein was known prior to the town's application for title and prior to subsequent discoveries.<sup>9</sup> However, the test normally announced has been that in order to except mineral lands from the operation of a town-site patent, it is not sufficient that the lands do in fact contain minerals, or even valuable minerals, when the town-site patent takes effect, but they must at that time be known to contain minerals of such extent and value as to justify expenditures for the purpose of extracting them. If the lands are not known at the time to be so valuable for mining purposes, the fact that they have once been valuable, or are afterwards discovered to be still valuable for such purposes, does not defeat or impair the title of persons claiming under the townsite patent.<sup>10</sup>

The townsite laws resulted in a concept that is to be considered unique in the administration of the early public land

<sup>7</sup> *Id.* at 526; also see *Tombstone* App 308, 44 Pac 69; see *Golden v. Townsite Cases* (1887) 2 Ariz 272, 15 Pac 26; 2 Ariz 442, 17 Pac 72. *Murphy* (1909) 31 Nev 395, 103 Pac 394.

<sup>8</sup> See *Brady's Mortgage v. Harris* (1900) 39 LD 426.

<sup>9</sup> *Mayle v. Bullene* (1896) 7 Colo

<sup>10</sup> *Dower v. Richards* (1894) 151 US 658; *Bonner v. Meikle*, *supra* N. 5; *Clark* (1928) 52 LD 426.

laws dependent upon classification of lands as mineral or non-mineral. The issuance of a townsite patent is not considered an adjudication that the lands embraced therein are non-mineral in character. In protests by claimants of unpatented mining claims against townsite patents embracing their claims within the limits of the townsite, the position has been taken that under the townsite laws a patent cannot operate to convey title to any lands known to be valuable for minerals at the date of the entry or which are held under a possession recognized by law, and the presence of such minerals is not a bar to issuance of a townsite patent. In such a situation, jurisdiction remains in the land department over any known minerals so that it may subsequently issue a patent to a mining claim when it is determined that the mining claim was in fact based upon rights, present or prospective, possessory or otherwise, which were acquired under provisions of the mining laws at the time of allowing the townsite patent.<sup>11</sup> Such an approach is an application of the principle that mining location rights recognized by law cannot be subsequently conveyed by the United States, but are to be protected by recognizing that the townsite patent cannot affect them.<sup>12</sup> The land office has demanded the clearest proof that land within a townsite patent was known to be valuable for minerals at the time of the patent whenever the issue has been raised many years thereafter.<sup>13</sup>

As in the case of other entries of public lands, a townsite entry consists of several formal steps spanning a period of time. The land department has expressed the view that the determination of any mineral claims encompassed within a townsite entry must be based upon facts existing as of the date of townsite entry.<sup>14</sup> The phrase "date of townsite entry" was held to refer to the date when final entry of the

<sup>11</sup> Golden Valley Mining Co. (1919) 47 LD 25; Brophy v. O'Hare (1906) 34 LD 596; Laland v. Townsite of Saltese (1903) 32 LD 211; Hulings v. Ward Townsite (1899) 29 LD 21; Pacific Slope Lode v. Butte Townsite (1897) 25 LD 518.

<sup>12</sup> See Butte City Smoke-House Lode Cases (1887) 6 Mont 397, 12 Pac 858; Talbott v. King (1886) 6 Mont 76, 9 Pac 434.

<sup>13</sup> Mill Side Lode (1910) 39 LD 356.

<sup>14</sup> Clark (1928) 52 LD 426.

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townsite is made and certificate of purchase is issued, or at least to the date when the townsite claimants have done everything required under the laws and regulations to entitle them to such certificate and nothing remains but to issue it. The latter position was adopted as being in accord with the general rule applicable to the vesting of rights in entrymen under the general land laws. The department recognized that the supreme courts of two of the mining states had previously held that the time for characterization of land within a claimed townsite was as of the time of the original application for entry.<sup>15</sup> However, the departmental decision clearly indicates that it would accept proof of the discovery of minerals at any time prior to the time all is done but the issuance of a final certificate as excluding the land from the townsite entry.

§ 3.19 ———. Entries on Mineral Lands. By the Act of March 3, 1891<sup>1</sup> provision was made for townsite entries by:

[I]ncorporated towns and cities on the mineral lands of the United States, but no title shall be acquired by such towns or cities to any vein of gold, silver, cinnabar, copper, or lead, or to any valid mining claim or possession held under the law . . . the possessor of such mineral vein may enter and receive patent for such mineral vein, and the surface ground appertaining thereto: *provided*, that no entry shall be made by such mineral-vein claimant for surface ground where the owner or occupier of the surface ground shall have had possession of the same before the inception of the title of the mineral-vein applicant.

The different features involved in the legislation are the limitation of the act to incorporated cities or towns, the authorization of entries on mineral lands, and specific authority for the issuance of a patent to veins of minerals excepted from the operation of the townsite patent. However, such legislative provisions had little effect upon the general rules applied to townsite entries under the earlier acts. It had been

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<sup>15</sup> See *Clark v. Jones* (1926) 116 § 3.19 '26 Stat 1101 (1891), 43 Ore 94, 240 Pac 551; *Lockwitz v. Larson* (1898) 16 Utah 275, 52 Pac 279. USC § 728 (1958).



was sold to the plaintiff at the same time, and, so far as we can see, his possession of her during the whole six months was open and unequivocal, carrying with it the usual marks and indications of ownership. It was such as to give evidence to the world of the claims of the new owner. It was an open, visible change manifested by such outward signs as rendered it evident that the possession of the vendor had wholly ceased. *Stevens v. Irwin*, 15 Cal. 506; *Cahoon v. Marshall*, 25 Cal. 201. And it was not necessary that the plaintiff's possession of the mare be continued indefinitely. It was said by REDFIELD, J., in delivering the opinion of the supreme court of Vermont in *Dewey v. Thrall*, 13 Vt. 284: "After a sale of personal chattels has become perfected by such a visible, notorious, and continued change of possession, that the creditors of the vendor may be presumed to have notice of it, the vendee may lend or let, or employ the vendor to sell, or perform any other service about the thing with the same safety he may a stranger." So in *Stevens v. Irwin*, *supra*, this court said: "This possession must be continuous—not taken to be surrendered back again—not formal, but substantial. But it need not, necessarily, continue indefinitely, when it is *bona fide* and openly taken, and is kept for such a length of time as to give general advertisement to the *status* of the property, and the claim to it by the vendee." See, also, *Godchaux v. Mulford*, 26 Cal. 316. In the light of these decisions we cannot say that the plaintiff's possession of the mare was not sufficient, the court below having found that it was, and we think, therefore, that the judgment and order should be affirmed.

We concur: FOOTE, C.; HAYNE, C.

By THE COURT. For the reasons given in the foregoing opinion the judgment and order are affirmed.

#### THE TOMBSTONE TOWN-SITE CASES.

REILLY v. BLACKMORE. SAME v. BERRY. MOUNTAIN MAID MIN. Co. v. REILLY.

(Supreme Court of Arizona. September 1, 1887.)

1. PUBLIC LANDS—TOWN-SITE—MINING CLAIM—CONFLICTING PATENTS.  
Where a patent to a town-site and a patent to a mining claim conflict, that one will be sustained which first vests the right.
2. SAME.  
A town-site patent does not vest a right in lands known at the time to be mineral lands.
3. SAME.  
A town-site patent will not be defeated by discovery of mineral and location of mineral lands as such after the town-site patent.
4. SAME.  
The right to mineral lands vests at the time of a valid location. A location void for uncertainty prior to a town-site afterwards amended and made the basis of a patent will not defeat a town-site patent prior to such amendment.
5. MINING CLAIM—LOCATION—EFFECT OF PATENT.  
The granting of a patent is *res adjudicata* that lands were mineral lands at the time of location, and known to be such.
6. SAME—EVIDENCE.  
A location uncertain as to lands claimed, unaided by proof of monuments, possession, or working, cannot be evidence that lands were then known to be mineral lands.

(Syllabus by the Court.)

Appeal from district court, Cochise county; BARNES, Judge.  
Thos. Mitchell and Goodrich & Smith, for appellees. Geo. G. Berry, for appellants.



BARNES, J. The question presented in these cases is between a title derived from the Tombstone town-site patent, and one derived from a patent to the Mountain Maid Mining Company. The date of entry of the town-site, is April 9, 1880. The mining patent was dated August 15, 1882. February 25, 1879, there was filed a notice of location of the Mountain Maid mine. This notice is so uncertain that the land claimed cannot be identified. It is aided by no evidence whatever. November 20, 1880, the record was amended, and on August 15, 1882, a patent was issued to the land described in the amendment. This amendment was after the entry of the town-site. The mining title cannot be superior to the town-site title, unless the location, earlier than the town-site, be held to have fixed the title. A location of a mining claim, to fix the title as against after acquired rights by entry and patent, should be sufficiently clear to designate the ground claimed, and should be marked on the ground by monuments, showing the extent of possession. If the location on its face be uncertain, the uncertainty could be aided by evidence of the possession, or of monuments; but a location notice, on its face uncertain and without evidence of what land was occupied, cannot be evidence for any purpose. An amendment afterwards made, describing different land or making certain what was uncertain, cannot revert back to the original defective location. The entry of the town-site intervening after the first location and before the amendment must be prior in right as it is prior in time. If this were not so, it would open the door to great wrongs. A person might locate a mining claim, and then, by doing \$100 worth of work annually, lie still and permit title to be acquired from a town-site, buildings erected, thousands of dollars spent, and then apply for a patent, amend his record, and swing his claim around, so as to include the most valuable improvements, and hold the same under his mining claim; and there is no limit to the time he might wait. It would be monstrous to give that construction to the mining act of 1872 and the town-site act as would make such a result possible. A town-site entry upon land not known to be mineral land at the time of entry is prior to any after acquired mining claim. *Deffebach v. Hawke*, 115 U. S. 392; 6 Sup. Ct. Rep. 95.

We have reviewed with care the *Butte City Smoke-House Lode Cases*, 12 Pac. Rep. 858, and *King v. Thomas*, Id. 865. We concur with these cases in holding that all mines, mining claims, and possessions held under existing laws are excluded from the terms of a town-site patent; that mining claims, located before the town-site entry, are paramount thereto. We also go so far as to say that lands upon which is any mine of gold, silver, cinnabar, or copper, or known to be such mineral lands, at the time of a town-site entry, are not included in such town-site entry. If those cases are to be construed as holding that if minerals be discovered in lands conveyed by a town-site patent, after the patent, and located as such, that such after-discovery and location become paramount to a town-site patent, and take such lands out from the operation of such patent, we do not concur with them. The question does not arise in the *Smoke-House Cases*. In the *King Case* there is some doubt as to whether this question is passed upon. [The court in that case properly held that whether the lands were as a matter of fact mineral lands is *res adjudicata* by the patent to the mining claim.] They held that the Silver King was a valid mining claim at the time of the issuing of the town-site patent. This view of these cases will meet with approbation, and clearly states the law as we understand it. The case before us, however, seeks to apply the principle to an invalid mining claim, prior to the town-site patent, and without proof that any mine existed prior to the town-site entry, or that the lands were known to contain mineral veins or deposits, or that there was any possession under existing laws. This is a very different question. To hold that every purchaser of a lot under a town-site patent, who erects valuable improvements thereon, ever after rests upon the precarious chance of having his title

and possession defeated by the discovery of minerals sufficient in amount to sustain a mining claim, is further than we can go. The supreme court in the *Deffebach Case*, *supra*, seem to have had this in mind, and to have recoiled from it, and they go no further than to hold "that a title to known mineral land cannot be acquired under the town-site laws."

We therefore hold that the Tombstone town-site patent is paramount to the patent to the Mountain Maid mine. The judgments are affirmed.

WRIGHT, C. J., concurs.

O'DOHERTY v. TOOLE.

(*Supreme Court of Arizona*. September 25, 1887.)

1. FRAUDULENT CONVEYANCES—BETWEEN HUSBAND AND WIFE.

Real estate, conveyed by a husband to his wife, where the deed has not been recorded, and where husband openly has charge of and use of property conveyed, where it is listed for taxation in his name, and he pays taxes on the same, may be reached to satisfy a judgment against the husband.

2. LIMITATION OF ACTIONS—CLAIMS AGAINST AN ESTATE.

The statute barring all claims against an estate, if not presented within 10 months, does not bar action to subject said property to a judgment.

3. EXECUTORS AND ADMINISTRATORS—ACTION AGAINST ESTATE—PARTIES.

The administrator is not a necessary party to such an action.

(*Syllabus by the Court.*)

Appeal from county court of Pima; GREGG, County Judge.

J. A. Anderson, for appellant. Hereford & Lovell, for respondent.

WRIGHT, C. J. The main questions raised by the record in this case are: (1) Had the appellant and creditor, under the general law, the right to pursue the property fraudulently conveyed by the debtor, in the hands of the vendee, without joining the executor of the deceased debtor? (2) If so, has the law of this territory contravened that right, or by it is the debt barred and the right lost, because the claim was not presented for allowance in the probate court within the 10 months allowed by said law?

John O'Doherty, the appellant and creditor, had recovered judgment against James H. Toole, the husband of respondent, Louisa M. Toole, and one Hudson, during the life-time of said Toole, in the district court of Pima county, for about \$3,800. The exact date of the judgment was May 29, 1884; the appellant having been a creditor of the said Toole some months prior to the date of the judgment. On the first day of November, 1882, Toole executed a deed to the respondent, his wife, whereby he deeded to her certain premises in the city of Tucson, worth at the time probably \$15,000. This was a voluntary conveyance, made in consideration of one dollar and love and affection; and the judge below, in his first conclusion of law, finds that the deed was fraudulent and void as to appellant, but that it vested a good title in respondent, as to said Toole, his heirs, devisees, and assigns; and, in his second conclusion of law, he finds that respondent was and is estopped from denying that said deed was and is void as to the creditors of the said Toole.

These findings or conclusions of law were undoubtedly correct. The evidence showed that, after Toole had executed this deed to his wife, the respondent, she and he suffered it to lie in a drawer, along with other papers of his, for nearly two years; that it was not recorded until two months after the assignment, in May, 1884; that during all that time the public, including appellant, had no intimation whatever that the property had been deeded by Toole to his wife; that, on the contrary, Toole gave the property in to the assessor as his, during the years 1883 and 1884, and paid the taxes thereon during those years; that he continued to act towards the property in every respect



5404 199

18395

01 DEC

UNION TITLE COMPANY

STATE OF ARIZONA, County of Maricopa, ss.

I do hereby certify that the within instrument was filed and recorded at request of

5404 199

JAN 29 1965 9 22 AM

M. D. Baird 5104

Records of Maricopa County, Arizona.

WITNESS my hand and official seal the day and year first above written

CLIFFORD H. WARD

Notary Public

By *Indenter* Deputy

When recorded, mail to:

ED MOONJIAN  
2025 East Campbell (Apt. 11)  
Phoenix, Arizona

*No Moonjian  
in the phone  
book*

Quit Claim Deed  
(JOINT TENANCY)

1.75

Policy No. 28740

For the consideration of Ten Dollars, and other valuable considerations, I or we,

D. D. BAIRD, a single man

hereby quit-claim to

ED MOONJIAN and NICOLE L. MOONJIAN, his wife

not as tenants in common and not as community property estate, but as joint tenants with right of survivorship the following described property situated in the County of Maricopa, State of Arizona:

An undivided 12 1/2% interest in and to those certain portions of the plat of VULTURE CITY described as follows:

PARCEL "A": That portion of the plat of Vulture City lying South of the center line of Yuma Street and West of the center line of Hayes Street;

PARCEL "B": That portion lying South of the center line of Prescott Street and East of the center line of the North-South alleys through Blocks 39 and 46 and West of the center line of Hancock Street;

PARCEL "C": That part North of the center line of Phoenix Street and East of the center line of Hancock Street;

ALL according to said plat of VULTURE CITY recorded September 21, 1880, and indexed in Book 1 of Maps, page 2, records of Maricopa County, Arizona.

(Consideration for this deed is less than \$100; therefore, no Internal Revenue Stamps are required)

The grantee by signing the acceptance below evidence their intention to accept and premises as joint tenants with the right of survivorship and not as community property or as tenants in common.

Dated this 29 day of JANUARY 1965

ACCEPTED AND APPROVED

*Ed Moonjian*  
*Nicole L. Moonjian*  
Grantors

*D. D. Baird*

STATE OF ARIZONA

County of Maricopa

Notary Public

This instrument was acknowledged before me this 29 day of January 1965

D. D. BAIRD, a single man

My commission expires

Notary Public

STATE OF ARIZONA

County of Maricopa

Notary Public

This instrument was acknowledged before me this 29 day of January 1965

ED MOONJIAN and NICOLE L. MOONJIAN, his wife

My commission expires

Notary Public

5414 210

18396

0100

STATE OF ARIZONA, County of Maricopa, ss

I do hereby certify that the within instrument was filed and recorded at request of

5414 210 - JAN 22 1935 - 9:00 AM

at Records of Maricopa County, Arizona

UNION TITLE COMPANY

M. Barker 5404

WITNESS my hand and official seal the day and year first above written

CLIFFORD H. WARD

Notary Public for Maricopa County, Arizona

When recorded, mail to

JOHN C. McCLAMROCH, JR.  
5515 E. Mockingbird Lane  
Scottsdale, Arizona

By *Indenter*

Deputy

Quit Claim Deed  
(JOINT TENANCY)

1-75

Policy No. 28740

For the consideration of Ten Dollars, and other valuable considerations, I or we,

D. D. BLAIR, a single man

hereby quit-claim to

JOHN C. McCLAMROCH, JR. and DONNA B. McCLAMROCH, his wife  
not as tenants in common and not as community property estate but as joint tenants with right of survivorship the following  
described property situated in the County of Maricopa, State of Arizona:

An undivided 12 1/2% interest in and to those certain portions of the plat  
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Blocks 39 and 46 and West of the center line of Hancock Street;

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and indexed in Book 1 of Maps, page 2, records of Maricopa County, Arizona.

(Consideration for this deed is less than \$100; therefore, no Internal  
Revenue Stamps are required)

January 19 1935  
ACCEPTED AND APPROVED

1935

*D.D. Blair*

Grantee

This instrument was acknowledged before me this 19th day of January, 1935, by D. D. BLAIR, and DONNA B. McCLAMROCH, his wife.

5404 no 195

01 DEED 18394

STATE OF ARIZONA, County of Maricopa, ss

I do hereby certify that the within instrument was filed and recorded at request of

5404 no 195

JAN 29 1965 - 2 00 AM

Page

Records of Maricopa County, Arizona

UNION TITLE COMPANY

M. Darter

WITNESS my hand and official seal the day and year first above written

CLIFFORD H. WARD

Notary Public, Maricopa County, Recorder

When recorded mail to

BERNARD MILLER  
2725 East Flower  
Phoenix, Arizona 85016

By: *Am. Union*

Deputy

**Quit Claim Deed**  
(JOINT TENANCY)

1-75

*In book  
same address  
956-4973*

Policy No. 28740

For the consideration of Ten Dollars, and other valuable considerations, I or we

D. D. BLAIR, a single man  
hereby quit-claim to

BERNARD MILLER and RUTH MILLER, his wife  
not as tenants in common and not as community property estate, but as joint tenants with right of survivorship, the following  
described property situated in the County of Maricopa, State of Arizona

An undivided 25% interest in and to those certain portions of the plat  
of VULTURE CITY described as follows:

PARCEL "A": That portion of the plat of Vulture City lying South of the  
center line of Yuma Street and West of the center line of Hayes Street;

PARCEL "B": That portion lying South of the center line of Prescott  
Street and East of the center line of the North-South alleys through  
Blocks 39 and 46 and West of the center line of Hancock Street;

PARCEL "C": That part North of the center line of Phoenix Street and  
East of the center line of Hancock Street;

ALL according to said plat of VULTURE CITY recorded September 21, 1980,  
and indexed in Book 1 of Maps, page 2, records of Maricopa County, Arizona.

(Consideration for this deed is less than \$100; therefore, no Internal  
Revenue Stamps are required)

The grantors by signing the acceptance below evidence their intention to acquire said premises as joint tenants with the right  
of survivorship and not as community property or as tenants in common

GRANTOR: D. D. BLAIR  
ACCEPTANCE AND APPROVAL

19 65

*B. D. Blair*

Grantors

STATE OF ARIZONA

This instrument was acknowledged before me  
this 29th day of January, 1965.

STATE OF ARIZONA

My Comm. Expires

My Commission Expires

STATE OF ARIZONA

COUNTY OF MARICOPA

I hereby certify that the within instrument was filed and recorded

IN DOCKET 15948

Dkt 15948 PG 848

848-849  
and indexed in DEEDS  
APR 9 - 1982 - 2 15DEED 0  
Fee No.

108473

Compared

Photostated

Fee: 5.00

at the request of

L.W. Beal

When recorded, mail to:

L.W. Beal  
P.O. Box 1853  
Wickenburg, Arizona 85358

Witness my hand and official seal.

BILL HENRY  
COUNTY RECORDER

R. B. Brounager

Deputy Recorder

**Warranty Deed**

For the consideration of Ten Dollars, and other valuable considerations, I or we, L. Wayne Beal, and M. Jeanette Beal, husband and wife, do hereby convey to V.M.P., INC., an Arizona corporation,

the following real property situated in Maricopa County, Arizona:

Refer to Exhibit "A" attached hereto and incorporated herein by this reference.

Subject to current taxes and other assessments, reservations in patents and all easements, rights of way, encumbrances, liens, covenants, conditions, restrictions, obligations and liabilities as may appear of record, the Grantor warrants the title against all persons whomsoever.

Dated this 14 day of March, 1982.

L. WAYNE BEAL

M. JEANETTE BEAL

STATE OF ARIZONA

County of Maricopa

ss.

This instrument was acknowledged before me, this 14 day of March, 1982, by L. Wayne Beal, and M. Jeanette Beal.

B. Brounager  
Notary Public

My commission will expire 2-3-86

STATE OF ARIZONA

County of

ss.

This instrument was acknowledged before me, this 14 day of March, 1982, by

Notary Public

My commission will expire



## DESCRIPTION

PARCEL NO. 1: HAMILTON, ASTOR, TALMAGE, CONKLING, CUSTER, SHERMAN, VAN BUREN, ELMORE, JANE ELMORE and SHERIDAN Lode Mining Claims, in Sections 35, and 36, Township 6 North, Range 6 West of the Gila and Salt River Base and Meridian, and in Section 31, Township 6 North, Range 5 West of the Gila and Salt River Base and Meridian in the Vulture Mining District, being shown on Mineral Survey No. 2511 on file in the Bureau of Land Management as granted by Patent recorded July 1, 1924 in Book 186 of Deeds, page 259, records of Maricopa County, Arizona.

PARCEL NO. 2: PIT GOLD Lode Mining Claim in the Southeast quarter of Section 36, Township 6 North, Range 6 West of the Gila and Salt River Base and Meridian in the Vulture Mining District, being shown on Mineral Survey No. 2522 on file in the Bureau of Land Management as granted by Patent recorded July 1, 1924 in Book 186 of Deeds, page 257, records of Maricopa County, Arizona.

PARCEL NO. 3: VULTURE EXTENSION Lode Mining Claim in the South half of the Northwest quarter of Section 31, Township 6 North, Range 5 West of the Gila and Salt River Base and Meridian in the Vulture Mining District, being shown on Mineral Survey No. 3146, on file in the Bureau of Land Management as granted by Patent recorded November 17, 1924 in Book 187 of Deeds, page 441, records of Maricopa County, Arizona.

PARCEL NO. 3-A:

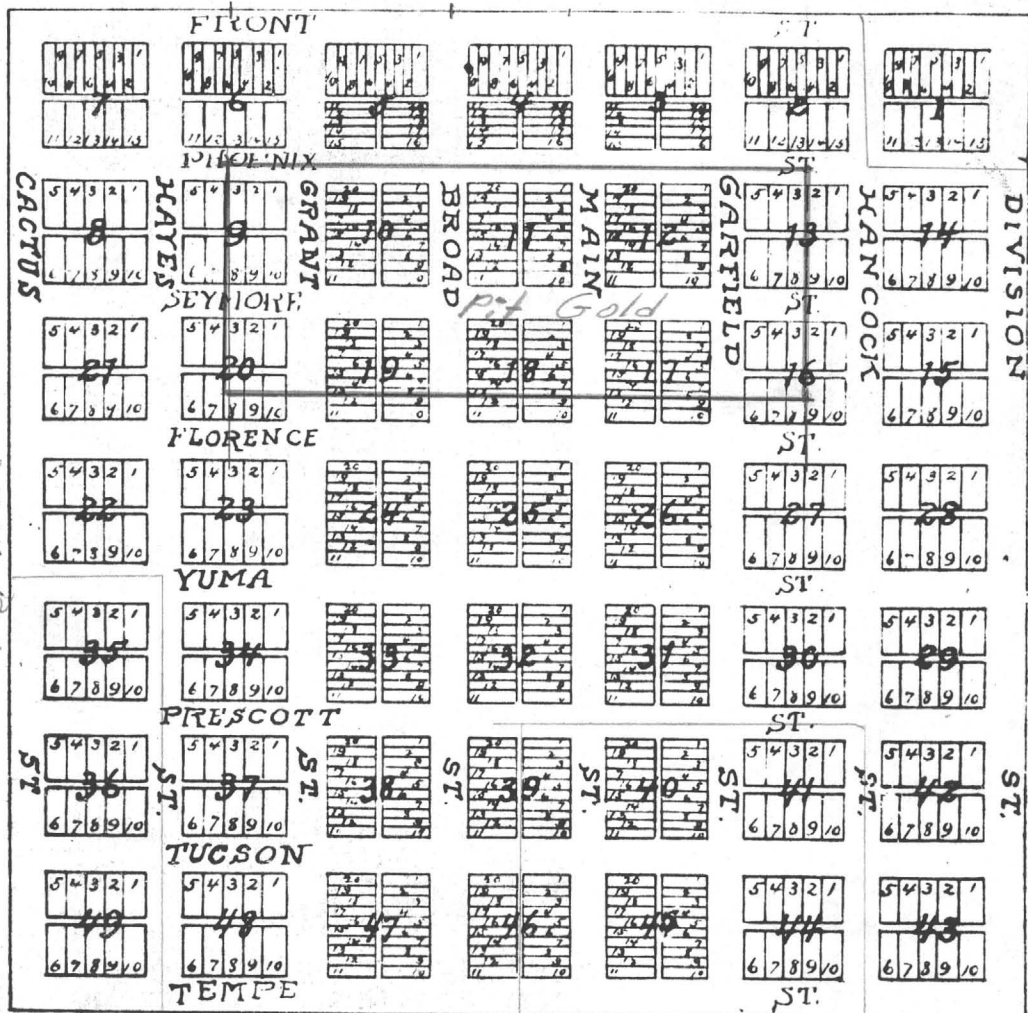
GOLD NUGGET LODE & CANON CITY LODE #1 Mining claim, in Vulture District in Section Sixteen (16), Township Six (6) North, Range Five (5) West, being shown on Mineral Survey No. 1797, on file in the Bureau of Land Management, as granted by Patent recorded February 21, 1974, in Docket 10526, page 323, records of Maricopa County, Arizona.

Pit Gold Claim M.S. 2522

patent recorded 7/1/24

305  
305

Needed: blocks 2, 3, 4



**EXPLANATORY**

Townsite 1/2 mile (6640ft) Square

Including 160 Acres -

All Streets are 90ft wide.

Except MAIN, BROAD, FLORENCE

and YUMA which are 105.

All Alleyways are 20ft. wide.

Interior Business Lots are 25x125ft.

Corner

in blocks 3, 4 and 5 which are 25ft. x125ft.

Interior Resident lots are 50ft. x125ft.

Corner

Plat of

**VULTURE CITY**

Maricopa Co. ARIZONA

Surveyed by

Herbert R. Patrick.

SURVEYOR.

1" = 500'  
3.673 deg/Block

(When?)

Book 1 Maps, page 2.

THE VALLEY ABSTRACT CO. PHOENIX ARIZONA.

# IRON KING ASSAY INC.

Page 1

28-Oct-86

LAB JOB #: MSC01101

Client name: DMEA Ltd.

No. Samples: 6  
Date Received: 10-22-86  
Submitted by: Don White

Billing address: 7340 E. Shoeman Ln.  
Suite 111-B-E  
Scottsdale, AZ 85251

Phone number: 778-3140

INVOICE ATTACHED

## ANALYTICAL REPORT

Client ID	Lab ID	FA/AA Au oz/ton
MSC01101		
M 101	1101- 1	0.003 M1 54-58 .003 <sup>✓</sup>
M 102	1101- 2	0.115 M1 61-62 .096 <sup>✓</sup>
M 103	1101- 3	3.190 M2 11-16 2.19 DMEA LTD.
M 104	1101- 4	0.012 M2 16-21 .006 <sup>✓</sup> OCT 29 1986
M 105	1101- 5	0.031 M3 24-27 .030 <sup>✓</sup> RECEIVED
M 106	1101- 6	0.002 M3 27-31 <.001



# Vulture Check Arrays

call on  
Friday  
622-4836

October, 1986

				Gold Array (0.2%)		Δ	ppm			
I.K. #		Hole	Footage	I K	Skyline					
1	MSC 1048	4	H-41	15-20	.009	.008		.26	.010	
2		5		20-25	.081	.082	+1.25%	2.8	.085	
3		6		25-30	.052	.038	-27%	1.3	.040	
4		7		30-35	.321			>10	.325	
5		8		35-40	.017	.015		.50	.015	
6		9		40-45	.003	.004		.14	.005	
7	MSC 1055	22	H-47	120-125	.004	.004		.15	.005	
8		23		125-130	.005	.004		.12	<.005	
9		24		130-135	.004	.003		.09	<.005	
10		25		135-140	.016	.018		.63	.020	
11	MSC 1056	28	H-48	150-155	.013	.013		.46	.015	
12		29		155-160	.015	.012		.40	.010	
13		30		170-175	.180	.140	-22%	4.8	.135	
14		31		175-180	.002	.002		.05	<.005	
15	MSC 1057	13	H-49	70-75	.009	.009		.32	.010	
16		14		75-80	.220	.239	+8.6%	8.2	.230	
17		15		80-85	.280	.277		9.5	.285	
18		16		85-90	.085	.143	+68%	4.9	.145	
19	MSC 1072	4	H-54	15-20	.100	.134	+34%	4.6	.135	
20		5		20-25	.185	.225	+22%	7.7	.190	
21	MSC 1080	5	M-1	34-39	.004	.003		.09	<.005	
22		6		39-43	.049	.050		1.7	.050	
23		7		43-46	.073	.061	-16.4%	2.1	.060	
24		19		80-84	.050	.058	+16%	2.0	.060	
25		20		84-85	.099	.088	-11.1%	3.0	.085	
26	MSC 1081	1	M-2	11-16	2.190			>10	2.400	
		2		16-21	.006	.006		.22	.005	
28		8		50-54	.080	.085	+6.25%	2.9	.085	
29	MSC 1082	2	M-3	19-24	.003	.038	+267%	1.3	.040	
30		3		24-27	.030	.006		.21	.005	

Total of 20 Reverse circ. samples and 10 core chip samples or 30 gold check arrays  
0.044 0.044



# ORDER FOR ANALYTICAL SERVICES

Samples Sent to:

## SKYLINE LABS, INC.

P.O. BOX 50106 • 1700 WEST GRANT ROAD  
TUCSON, ARIZONA 85703  
(602) 622-4836

(Report and invoice in duplicate will be sent to address below unless otherwise instructed)

Address Report To:

Don White  
521 East Willis St.  
Prescott, AZ  
86301

Tel. 778-3140

P.O. NO.:

SHIPMENT NO.:

DATE SHIPPED: Tuesday, Oct 14, 1986

SHIPPED VIA: U.P.S. (insured)

NO. OF CARTONS: One (1)

NO. OF SAMPLES: Thirty (30)

(Information above helps us trace lost shipments)

Send Invoice To: Ben F. Dickerson III  
DME A Ltd.  
7240 East Sherman Ln.  
Suite 111 - R - (E)  
Scottsdale, AZ 85251

Send Copy of Report To: Same

LIST SAMPLE NOS.	DESCRIBE MATERIAL	LIST ELEMENTS TO BE DETERMINED (Give anticipated range of values, if possible) Describe any special sample preparation procedures desired.	INDICATE METHOD OF ANALYSIS*	✓ IF 31 - ELEMENT EMISSION SPEC SCAN DESIRED
MSC 1048	4 thru 9	(6)	G	
MSC 1055	22 thru 25	(4)		
MSC 1056	28 thru 31	(4)		
MSC 1057	13 thru 16	(4)		
MSC 1072	4 and 5	(2)		
MSC 1080	5, 6, 7, 19, 20	(5)		
MSC 1081	1, 2, 8	(3)		
MSC 1082	2, 3	(2)		
<p><u>Gold assay</u> <u>by Fire/AA</u> <u>All are using one</u> <u>pulps assay ton</u> <u>and being</u> <u>care to homogenize</u> <u>each pulp prior to</u> <u>taking your split</u> <u>for assay</u></p> <p><u>If any values exceed the geochem (AA) range, go ahead with straight assay</u></p>				

PAYMENT FOR SERVICES REQUESTED MUST ACCOMPANY ORDER UNLESS CREDIT ARRANGED

Signature of person authorizing work: Don White

(Use Continuation Sheet If Necessary)

### INSTRUCTIONS

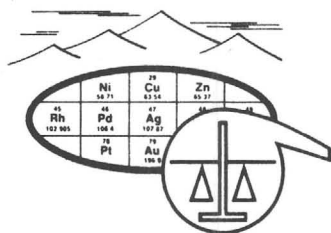
\*METHOD OF ANALYSIS: G-Geochem, Q-Quantitative or Routine Assay  
W-Wet Assay, F-Fire Assay

†SAMPLE STORAGE: Pulps stored 90 days pending instructions, bulk rejects stored 30 days pending instructions.

Enclose yellow original with samples, send white copy by mail, retain pink copy. White copy will be returned to shipper as an acknowledgement that shipment has been received.

INDICATE DESIRED DISPOSITION OF SAMPLES AFTER ANALYSIS	Bulk Rejects	Pulp
Return at customer's expense via: <u>None</u> (X)	<u>None</u>	<u>Save</u>
Store temporarily pending instructions†		
Discard immediately		

SENDER COPY



# SKYLINE LABS, INC.

1775 W. Sahuaro Dr. • P.O. Box 50106  
Tucson, Arizona 85703  
(602) 622-4836

## REPORT OF ANALYSIS

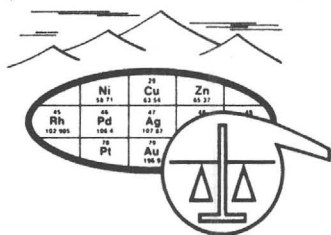
JOB NO. UQX 045  
October 29, 1986  
MSC-1048-04 TO  
MSC-1082-03  
PAGE 1 OF 2

A.F. BUDGE (MINING) LIMITED  
Attn: Mr. Ben F. Dickerson III  
DMEA Ltd.  
7340 E. Shoeman Lane, 111-B (E)  
Scottsdale, Arizona 85251

### Analysis of 30 Pulp Samples

		FIRE ASSAY
		Au*
ITEM	SAMPLE NUMBER	(oz/t)
1	MSC-1048-04	.010
2	MSC-1048-05	.085
3	MSC-1048-06	.040
4	MSC-1048-07	.325
5	MSC-1048-08	.015
6	MSC-1048-09	.005
7	MSC-1055-22	.005
8	MSC-1055-23	<.005
9	MSC-1055-24	<.005
10	MSC-1055-25	.020
11	MSC-1056-28	.015
12	MSC-1056-29	.010
13	MSC-1056-30	.135
14	MSC-1056-31	<.005
15	MSC-1057-13	.010
16	MSC-1057-14	.230
17	MSC-1057-15	.285
18	MSC-1057-16	.145
19	MSC-1072-04	.135
20	MSC-1072-05	.190
21	MSC-1080-05	<.005
22	MSC-1080-06	.050
23	MSC-1080-07	.060
24	MSC-1080-19	.060
25	MSC-1080-20	.085





# SKYLINE LABS, INC.

1775 W. Sahuaro Dr. • P.O. Box 50106  
Tucson, Arizona 85703  
(602) 622-4836

JOB NO. UQX 045  
October 29, 1986  
PAGE 2 OF 2

FIRE ASSAY  
Au\*  
(oz/t)

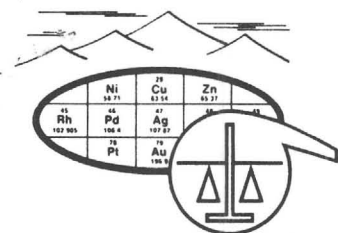
ITEM	SAMPLE NUMBER	
------	---------------	--

26	MSC-1081-01	2.400
27	MSC-1081-02	.005
28	MSC-1081-08	.085
29	MSC-1082-02	.040
30	MSC-1082-03	.005

\*NOTE: Analysis based on a one assay-ton sample.

cc: Mr. Don White  
521 East Willis St.  
Prescott, AZ 85301

REGISTERED ASSAYER  
CERTIFICATE NO.  
9425  
WILLIAM L.  
LEHMBECK  
SIGNED  
William L. Lehmbeck  
Manager  
10/29/86



# SKYLINE LABS, INC.

1775 W. Sahuaro Dr. • P.O. Box 50106  
Tucson, Arizona 85703  
(602) 622-4836

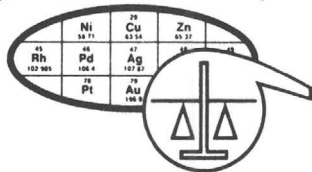
## REPORT OF ANALYSIS

JOB NO. UQX 045  
October 29, 1986  
MSC-1048-04 TO  
MSC-1082-03  
PAGE 1 OF 2

A.F. BUDGE (MINING) LIMITED  
Attn: Mr. Ben F. Dickerson III  
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### Analysis of 30 Pulp Samples

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5	MSC-1048-08	.015
6	MSC-1048-09	.005
7	MSC-1055-22	.005
8	MSC-1055-23	<.005
9	MSC-1055-24	<.005
10	MSC-1055-25	.020
11	MSC-1056-28	.015
12	MSC-1056-29	.010
13	MSC-1056-30	.135
14	MSC-1056-31	<.005
15	MSC-1057-13	.010
16	MSC-1057-14	.230
17	MSC-1057-15	.285
18	MSC-1057-16	.145
19	MSC-1072-04	.135
20	MSC-1072-05	.190
21	MSC-1080-05	<.005
22	MSC-1080-06	.050
23	MSC-1080-07	.060
24	MSC-1080-19	.060
25	MSC-1080-20	.085



# SKYLINE LABS, INC.

1775 W. Sahuaro Dr. • P.O. Box 50106  
Tucson, Arizona 85703  
(602) 622-4836

JOB NO. UQX 045  
October 29, 1986  
PAGE 2 OF 2

## FIRE ASSAY

ITEM SAMPLE NUMBER Au\*  
(oz/t)

26	MSC-1081-01	2.400
27	MSC-1081-02	.005
28	MSC-1081-08	.085
29	MSC-1082-02	.040
30	MSC-1082-03	.005

\*NOTE: Analysis based on a one assay-ton sample.

cc: Mr. Don White  
521 East Willis St.  
Prescott, AZ 85301

# Vulture Core for Metallurgical Tests

October, 1986

Hole No.	From	To	Interval (ft)	Rock Type	Assay oz/t	Weight (lbs)
M-1	39	43	4	h.w.	0.049	130.8
	43	46	3	qpi	0.073	98.1
	46	48	2	qpi	0.006	65.4
	48	50	2	qpi	0.028	65.4
	50	52	2	qpi	0.003	65.4
	52	54	2	qpi	0.053	65.4
	54	58	4	int.	0.003	130.8
	58	61	3	int.	0.006	98.1
	61	62	1	int.	0.096	32.7
	62	65	3	int.	0.011	98.1
	65	70	5	int.	0.013	163.5
	70	75	5	int.	0.009	163.5
	75	80	5	qpi	0.042	163.5
	80	84	4	qpi	0.050	130.8
	84	85	1	f.w.	0.099	32.7
M-2	11	16	5	h.w.	2.190	163.5
	16	21	5	h.w.	0.006	163.5
	21	27	6	h.w.	0.005	196.3
	27	32	5	h.w.	0.015	163.5
	32	40	8	h.w.	0.012	261.7
	40	45	5	h.w.	0.026	163.5
	45	50	5	h.w.	0.012	163.5
	50	54	4	h.w.	0.080	130.8
	54	58	4	h.w.	0.017	130.8
	58	63	5	h.w.	0.009	163.5
M-3	63	68	5	h.w.	0.002	163.5
	14	19	5	f.w.	0.011	163.5
	19	24	5	f.w.	0.003	163.5
	24	27	3	f.w.	0.030	98.1
	27	31	4	f.w.	< 0.001	130.8
	32	37	5	f.w.	0.005	163.5
	37	41	4	f.w.	0.005	130.8

Notes - BFD October, 1984

Abstract of NEI report on Vulture Prospect by Michael Donnelly  
March 16, 1981

### Recommendations

High grade mineralization restricted to zones of quartz veining and adjacent wall rock. Gold in metasediments has direct correlation with intensity and extent of hydrothermal alteration; altered portions of quartz monzonite intrusive are weakly mineralized.

Best potential for gold mineralization on meaningful scale (NEI) is restricted to the intrusive body; veins near exhaustion and gold in metasediments very low grade.

IP survey revealed nothing much of value; one weak anomaly - maybe one drill hole on it.

Property of some interest, but high cost deal precludes work at this time. (Beal mentioned unfavorably.)

### Introduction - History

Nothing much new. Two north dipping, east raking ore shoots mined; west shoot to 600 feet; east shoot to approx. 1000 feet; one stope on west shoot, 100-ft. level was approx. 80 feet wide. Noranda had 6-month walk-on at \$1,000/month. In 1970, conducted a bulk sampling program. No results given.

Present work consisted of rock chip and channel samples; detailed mapping in pits; and 20 shallow rotary holes (24-36 feet deep).

### Regional Setting

Nothing new - Rehrig

### Geology - Mine Area

#### Protovolcanics & sediments

Series of metavolcanics, amphibolites, fine- to coarse-grained epiclastics, quartz-pebble conglomerate and some volcaniclastics; a small lens (?) of ferruginous chert located north of pits is only chemical sediment seen. Upper greenschist metamorphism (staurolite to chlorite); some folding and pegmatites (aplite?).

Metasediments = wackes, quartz-wackes, siltites and quartzites; quartz pebble lens (?) exposed in pits; 75 feet thick in east pit; matrix strongly foliated quartz-sericite.

No bed top indicators: rock probably equivalent to Spud Mountain, or, Iron King volcanics: approx. 1760 m.y. age

#### Quartz Monzonite Porphyry Intrusive

Least understood rock in the mine area; poorly exposed; multiple facies and alteration overprint make it difficult to map.

Field evidence indicates Qm brought hydrothermal alteration to metasediments along with aplite and pegmatites; latter two very subordinate to Qm porphyry. Outcrop extent of Qmp is poorly defined; that exposed west of mine area dips under gravels to south, east and west. Mapping does show intrusive persists for 1 mile to north and northwest.

Small exposure of Qmp in pits in central part of mine area represents thin knobby apophysis of main intrusive mass. Qmp not seen underground.

#### Tertiary

Mid-Tertiary volcanics mostly andesite lavas and tuffs with some intercalated epiclastic and volcanoclastic materials. A 1929 Mining Journal article suggests 500 to 600 feet of volcanics on east side of Schoolhouse Fault in drill holes. The only Tertiary unit in the mine area is porphyritic rhyolite dike exposed in southwesternmost pit.

#### Quartz veins - zones

Two large and several smaller veins provided most of historic production; sub-parallel, east-west, semi-conformable to enclosing metasediments. Two large veins tentatively traced through pit areas for 900 feet.

Veins mostly quartz-carbonate with a little pyrite, chalcopyrite and galena; native gold in flakes and leaves in association with basemetal rich portions.

All mineralized vein samples show multiple stages of brecciation, silicification and annealing in thin section and hand specimen. Silicified wall rock fragments occur in all veins and show ghost outlines; changes color of veins to blue-grey; may be a quartz vein breccia.

Absence of cross-cutting relationships prevents establishing relative ages among veins. Two barren vein events in mine area; 2-30 mm thick veins in metasediments.

#### Structure

Fabric dominated by west trending, north dipping metasediments; both Qmp intrusive and quartz veins are partly controlled by rock structural grain; 35 to 55 degree north dips.



Limited folding, northeast plunging isoclinal; wave length less than 1 meter in east wall of west pit.

#### Faulting

Property in broad zone of intense north-northwest to northwest trending normal faults - shaped Vulture range.

Three episodes can be seen

1) In mine area: east-west to west-northwest trending high angle normal faults, predating quartz veins. These provided avenue for hydrothermal alteration best seen in east pit.

2) Post mineral faulting a) low angle normal faults displace quartz veins in stair step fashion; b) north-northwest trending high angle faults with right lateral movement; age relationship between (a) and (b) not known.

3) North trending Basin and Range faulting; youngest episode in area. The Schoolhouse Fault exemplifies; may have 500 to 600 feet of vertical displacement.

#### Hydrothermal Alteration

All metasediments and Qpi rocks in mine area have been hydrothermally altered. Quartz-sericite-pyrite is characteristic, and is most advanced in the Qpi apophysis and in fine-grained clastic units.

As a rule, metasediment alteration is strongly developed adjacent to mineralized quartz veins; alteration decreases laterally for 10's of feet before it becomes indiscernable with greenschist metamorphism. However, this relationship breaks down when alteration is structurally controlled and alteration overlaps. Structurally controlled alteration best seen in southwestern corner of east pit.

Advance alteration in metasediments frequently obliterates all relict sedimentary features. Fine-grained clastics are most susceptible and are usually replaced by an auriferous qtz-ser-py association. Pyrite may form 1-5% volume of rock.

#### Mineralization

Rock chip geochem in conjunction with mapping shows that gold mineralization occurs in 2 distinct modes:

1) mineralized quartz veins and in their hydrothermally altered wall rocks;

2) disseminated gold mineralization in Qpi. Although copper, lead, zinc and silver are present in anomalous amounts, gold is the principal metal. Gold-silver ratios vary from 1:1 in weakly altered metasediments to 1:3.8 in strongly mineralized

quartz veins.

Base metal concentrations are low in all units, with the exception of high grade samples. Arsenic content is quite low, 15-20 ppm, and shows no relationship to gold. Gold vs. lithology plot clearly shows gold mineralization not confined to a single stratigraphic unit or rock type. Gold does vary with the intensity of alteration.

Gold mineralization in Qpi is poorly understood. Rock chip analyses suggest that Qpi in the immediate mine area is weakly gold bearing and there is correlation between intensity of alteration and gold content. Pyritic samples usually the most strongly mineralized. Gold values in Qpi range from a low of 0.04 ppm to a high of 2.2 ppm.

#### IP Survey

Designed to detect possible sulfide-rich portions of intrusive body and to delineate strongly altered metasediments (Mining Geophys. Surveys of Tucson); 4 north-south trending lines, 500-600 feet apart.

Result: weak amplitude IP response with associated weak resistivity signal on 2 adjacent lines. Anomaly described as an east-west trending dike-like response at, or near the surface that is associated with a relatively high resistivity rock of 500 ohmmeters vs. background of about 150. The significance of this resistivity contrast is not clear. Geophys. estimated that under ideal conditions the 20 ms anomaly may equate to 1% volume sulfides.

The centre of the anomaly lies between the west open pits and the central mine area and continues to the east where it is very weak on line 4, which is centered over old mine workings in central pit of mine area.

Cross sections B-B' and C-C' show the north dipping projection of the pyritic altered rocks and is a likely explanation for the IP response. Rocks (altered metasediments and Qpi) in this area contain 0.5 to 1.0% volume pyrite.

#### Discussion

This work has added a certain amount of understanding to the gold mineralization at Vulture:

1) altered metasediments in mine area are ubiquitously mineralized;

2) gold content of more than 1.7 ppm (0.05 oz/t) in metasediments generally restricted to zones of more intense alteration adjacent to quartz veins;

3) gold content fades laterally for 10's of feet before

falling to background levels;

4) gold mineralization is not associated with or restricted to a particular stratigraphic unit, including quartz pebble conglomerate;

5) preliminary evidence suggests that the altered portions of Qpi body are weakly mineralized.

Evidence gathered during this work suggests the altered Qpi body offers the best potential for moderate to large tonnage, low grade gold mineralization. This is little promise for discovery of high grade gold vein mineralization on a meaningful scale; gold bearing metasediments are low grade.

The auriferous character of the intrusive is poorly understood and geologic and geochemical data are restricted to scattered exposures. Although preliminary model suggests a strong pyrite-gold correlation, a sulfide-poor mineralized system cannot be ruled out.

Relative and absolute ages of veining and emplacement of Qpi body are uncertain. Based upon petrologic makeup of the intrusive and the apparently deep erosional level, an atypical Laramide age magmatic event is suggested. Field evidence indicates that mineralized veins are cogenetic with the Qpi intrusive mass and unrelated to Tertiary volcanics.

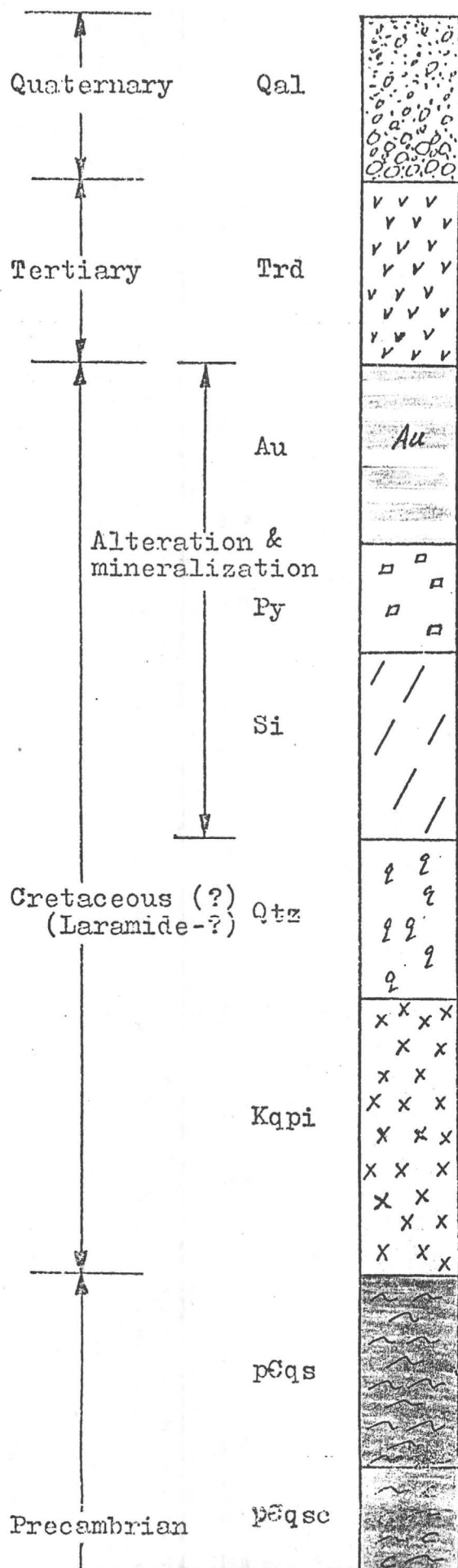
The absence of subsurface data in the west half of the mine area causes source of IP anomaly to be conjectural. From the geologic framework available now, several geologic explanations could be constructed. However, Gordon W's (geophys) interpretation that the anomaly is a near surface, north dipping, dike-like feature suggests that the subsurface projection of weakly to moderately altered intrusive and metasediments exposed on the surface can best account for the anomaly. The strength of the anomaly, in contrast to the weak response over vein mineralization on line 4 suggests that vein mineralization cannot be sole source. An attractive alternate explanation for the IP response is a sulfide poor, potentially mineralized, buried portion of the intrusive body.

- End -

# VULTURE MINE

## GRAPHIC DRILL LOG LEGEND

Compiled from rotary-reverse circulation chips



ALLUVIUM, colluvium; derived from the same rocks as in this sequence except for granodiorite from  $\geq 2$  miles NW. Varied but generally low degree of rounding, sorting, and stratification.

RHYOLITE DIKE; white, aphanitic, cross-cutting unit up to 15 feet in drill intercept thickness. Commonly contains black Mn dendrites on joint surfaces. Only known to occur in W portion of property near pits 3 and 4.

GOLD; noted in graphic log where significant assays are reported. Thought to be cogenetic with other Laramide (?) events like qtz-pphy intrusion and associated pyritization, silicification, and quartz veining (see below).

PYRITE; generally tiny ( $\leq 0.5$ mm) disseminated cubes. Often in qtz or intensely silicified rocks.

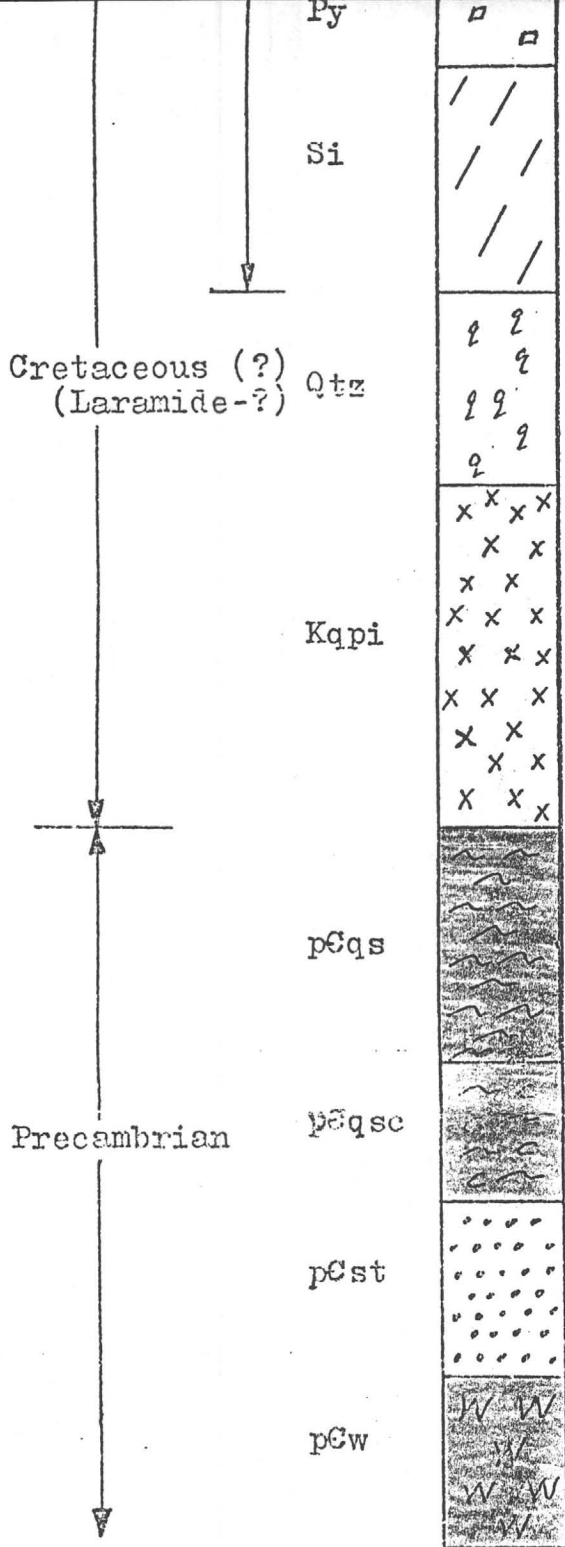
SILICIOUS ROCK; thought to be introduced silica related to the qtz-pphy intrusive. Often obscures foliation and sometimes prevents identity of the original rock type. Intense silicification yields an amorphous quartzite.

QUARTZ; only used to designate vein or bull quartz and discrete, visible qtz as in qtz-rich siltite and qtz-pphy intrusive. May harbor sulfides (py, gal) and native gold.

QUARTZ-PORPHYRY INTRUSIVE; med. to coarse-grained, altered (sericitized, pyritized silicified) granite to qtz-monzonite with qtz porphyroblasts (often up to 4mm dia). Quartz is typically a soft, pastel green (chloritic-?) or pale gray to milky. Emanates as an apophysis from stock to the W of pit 3. Generally semi-conformable, sill-like.

QUARTZ-SERICITE SCHIST; brown, gray, tan, or almost white, thin laminated, fine to med grained quartz and sericite. Often iron stained. Gradational to siltite or wacke (see below) and prone to silicification.

QUARTZ-SERICITE-CHLORITE SCHIST; same as above with the addition of a chlorite component (usually  $< 20\%$ ).



silicified rocks.

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QUARTZ-SERICITE-CHLORITE SCHIST; same as above with the addition of a chlorite component (usually <20%).

SILTITE; brown or tan to light gray, poorly foliated, very fine grained, meta-silt. Often a quartzite. Grades finer to quartz-sericite schist or coarser to wacke.

WACKE; brown, tan, or med to dark gray, poorly foliated, med grained, meta-wacke. Grades to siltite and qtz-ser schist. Prone to silicification.

Don White - Dec. 1984



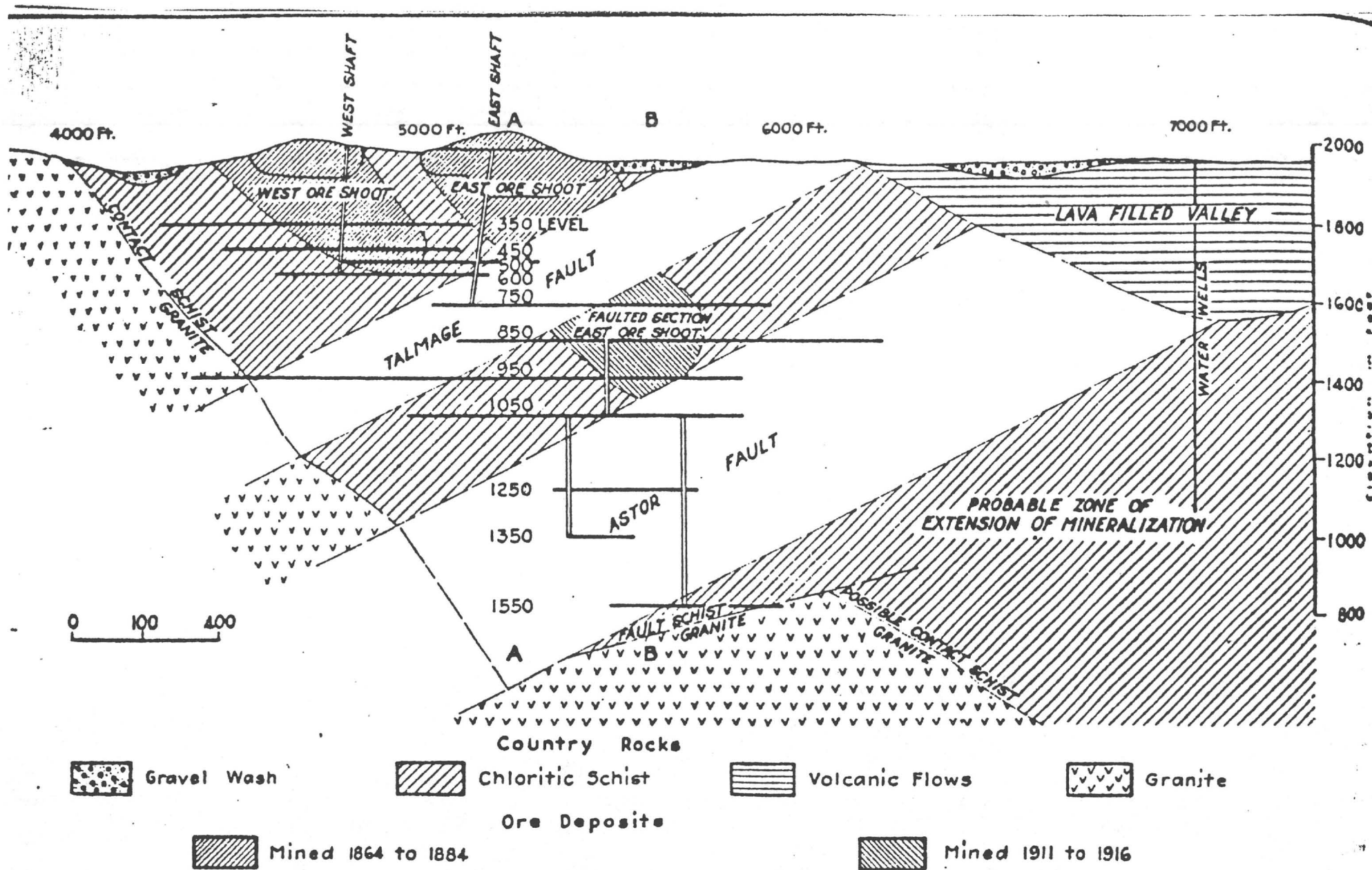


FIG. 1. LONGITUDINAL VERTICAL PROJECTION OF THE VULTURE MINE WORKINGS