Bureau of Mines Report of Investigations 5395



# ILMENITE AND OTHER BLACK-SAND MINERALS IN THE GOLD FORK PLACER DEPOSIT, VALLEY COUNTY, IDAHO

BY R. H. STORCH

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## UNITED STATES DEPARTMENT OF THE INTERIOR Fred A. Seaton, Secretary BUREAU OF MINES Marling J. Ankeny, Director

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by

### R. H. Storch 1/

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#### SUMMARY AND CONCLUSIONS

This report describes work done as part of a Bureau of Mines program to appraise the titanium-mineral resources of Idaho. The work included both field reconnaissance and a spot churn-drilling program. Much general information was available, as a result of earlier work done under the Western Radioactive Minerals Exploration Program conducted between 1950 and 1954 by the Bureau of Mines for the United States Atomic Energy Commission. This particular exploration provided data as to the quantity of ilmenite (TiFeO<sub>3</sub>) and accessory black-sand mineral concentrates in a number of placer deposits; however, at that time no tests were made to ascertain if the ilmenite was of marketable grade.

The spot churn drilling was begun in the Gold Fork area of Long Valley, Valley County, Idaho (fig. 1), in August 1956 and was completed during October of the same year. Eleven holes (designated as the GF series on fig. 2) 35 to 45 feet in depth, aggregating 425 feet, were completed during this period. Data from other exploration drilling in 1951 was used in this report. The 13 holes in the G series shown in figure 2 were drilled by the Bureau of Mines for the United States Atomic Energy Commission; the 7 holes of the DH series were drilled by the Bureau of Reclamation. The 20 holes, ranging from 16 to 140 feet in depth, totaled 1,625 feet. The combined number of all samples recovered was 422. Those taken in 1956 were composited for each drill hole and sent to the Bureau of Mines laboratory in Albany, Oreg., for analyses.

The areal extent of the property was not delineated entirely by drilling. Because of the small drilling unit, the CF series of holes in the central portion of the deposit did not test the black-sand values below a maximum depth of 45 feet, although previous drilling in the area indicated mineral values to depths exceeding 100 feet.

#### INTRODUCTION

Ilmenite (FeTiO3) is one of the heavy minerals that occur in the black sands found in varying amounts in most placer deposits.

Ilmenite has become of increasing economic importance as a raw material used in producing titanium pigments and titanium metal. This importance will increase because the less abundant titanium mineral, rutile (TiO2), which is preferred by most titanium-metal producers, may not be available in sufficient quantity to meet future demands.

In July 1956 the Bureau of Mines initiated a program to investigate deposits of titanium minerals in the United States and its possessions in order to provide sources of supply for the titanium industry. Idaho's titanium resources were studied as part of this program.

Accessory black-sand minerals occurring in the placers that might favorably affect development of the deposits were also included in this study. The more important of these minerals are monazite, thorite, uranothorite, columbite, and columbium-tantalum-bearing uranium minerals.

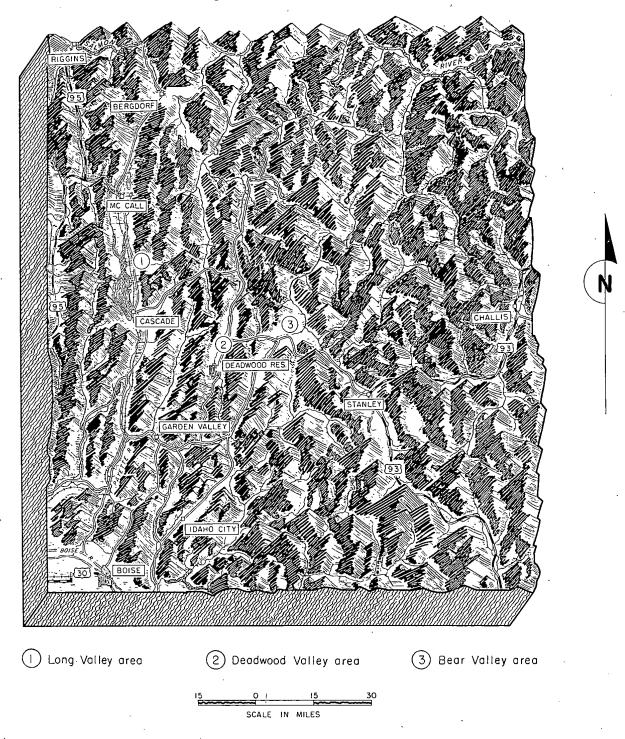


Figure 1. - Relief map of south central Idaho.

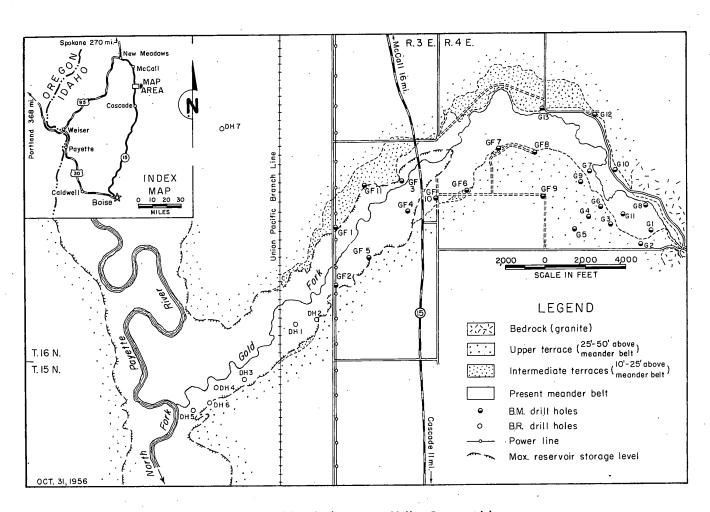


Figure 2. - Gold Fork placer area, Valley County, Idaho.

Work by the Federal Bureau of Mines for the United States Atomic Energy Commission between 1950 and 1954 2 3 4 5 6 7 8 9 10/ under the Western Radioactive Minerals Program, provided some information that is included in this report. The Bureau of Reclamation also did some exploration work, which was supervised by the Bureau of Mines, on the downstream portion of the deposit. These sample data were made available to the Bureau and provided additional information regarding the Gold Fork placer deposit.

#### ACKNOWLEDGMENTS

The writer wishes to express his appreciation to the following property owners for their cooperation and courtesies extended to the employees and staff of the Bureau:

George M. Loomis	Donnelly, Idaho
Floyd Loomis	Do.
G. E. Davis	Do.
G. W. Davis	Do.
Laurel Hansen	Do.

Harold D. Hess (supervising geologist), William A. Stickney (production metal-lurgist), and Peter Romans (supervising physicist), of the Albany, Oreg., laboratory, processed the black-sand concentrates and furnished analytical results.

The United States Atomic Energy Commission and the Bureau of Reclamation provided some exploration and sample data that were used in the report.

- Z/ Eilertsen, D. E., and Lamb, F. D., A Comprehensive Report of Exploration by the Bureau of Mines for Thorium and Radioactive Black-Mineral Deposits: Tech. Inf. Services, U. S. Dept. of Commerce, RME-3140, June 1956, 46 pp.
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- 4/ Kline, M. H., Carlson, E. J., and Storch, R. H., Big Creek Monazite Placers, Valley County, Idaho: Tech. Inf. Services, U. S. Dept. of Commerce, RME-3131, April 1951, 24 pp. Scott Valley and Horsethief Basin Monazite Placers, Valley County, Idaho: Tech. Inf. Services, U. S. Dept. of Commerce, RME-3133, June 1951, 22 pp.
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- 6/ Kline, M. H., and Carlson, E. J., Pearsol Creek Monazite Placer Area, Valley County, Idaho: Tech. Inf. Services, U. S. Dept. of Commerce, RME-3134, February 1954, 24 pp.
- 7/ Storch, R. H., and Robertson, A. F., Beaver Creek Monazite Placer Area, Valley County, Idaho: Tech. Inf. Services, U. S. Dept. of Commerce, RME-3132, June 1954, 15 pp.
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- 9/ Robertson, Almon F., and Storch, R. H., Camp Creek Radioactive Mineral Placer Area, Blaine and Camas Counties, Idaho: Tech. Inf. Services, U. S. Dept. of Commerce, RME-3136, June 1955, 27 pp.
- 10/ Robertson, A. F., and Storch, R. H., Rock Creek Radioactive Mineral Placer Area, Blaine County, Idaho: Tech. Inf. Services, U. S. Dept. of Commerce, RME-3139, June 1955, 25 pp.

#### HISTORY

There is no record of mineral production from the Gold Fork placer deposit in Long Valley. Near the headwaters of the North Fork of the Gold Fork River, hydraulicking and hand mining for placer gold were done before 1900. At the confluence of the North Fork of the Gold Fork and the Gold Fork, about 2 miles east of Long Valley, is Little Valley. J. C. Blood, in a private report dated 1917, wrote that extensive drilling and shaft testing in Little Valley indicated 48,000,000 cubic yards of material averaging 22 cents per cubic yard in gold. A small dredge was later installed, but the mining venture was unsuccessful.

In 1951 the Bureau of Mines did a limited amount of exploration in the Little Valley area, but the results indicated extremely low gold values and virtually no black-sand concentrates except on the rims in ancient stream gravels.

In 1951 the Cosumnes Dredging Co. of California drilled several holes on the north side of the Gold Fork deposit in Long Valley. Apparently the results were not encouraging, as the options to lease were dropped the same year.

Drilling by the Bureau of Mines in the Gold Fork area is discussed under the heading "Work by the Bureau of Mines."

#### LOCATION AND ACCESSIBILITY

The portion of the Gold Fork placer area that was investigated is in secs. 24, 25, 26, and 34, T. 16 N., R. 3 E., B. M., and sec. 30. T. 16 N., R. 4 E., B. M., Valley County, Idaho. The deposit is on the east side of Long Valley, about 11 miles north of Cascade, Idaho, and 100 miles north of Boise, the State capital (fig. 2).

The property is traversed by State Highway 15, a branch line of the Union Pacific Railroad, and a high-voltage transmission line owned by the Idaho Power Co. Secondary county and private roads permit reasonable access to most of the area except below the terraces bordering the present stream channel east of the maximum flowline of the Cascade reservoir. A limited amount of drilling west of the State highway is possible during stages of low water in the reservoir basin.

#### PHYSICAL FEATURES AND CLIMATE

The straight-line distance, between a point where the Gold Fork River emerges from a narrow canyon at the extreme eastern limit of the alluvial deposit and its confluence with the North Fork of the Payette River to the west, is about 5 miles (fig. 2). Gold Fork River flows through a trough or channel flanked by gravel terraces or benches ranging from 10 to more than 50 feet in height. The width of the trough is about 1,500 feet at the eastern end and more than 3,000 feet at the western extremity. The basin area that is bounded by gravel terraces forms part of the Cascade Reservoir area for almost 2 miles upstream on Gold Fork from its mouth. As a result, about 50 percent of the total basin area is under water several months each year. Upstream from the reservoir site virtially all the basin floor is swamp area covered with thick brush and a few scattered fir trees on higher mounds. The swamp conditions result from irrigation water that percolates into the areas of lower elevation from cultivated fields on the surrounding terraces. The irrigated benches are used for grazing stock and cultivating grain, seed potatoes, and clover seed.

The overburden of top soil ranges from less than 1 foot to 4 feet. The depth of the water table was found to be from a few inches to over 30 feet, depending upon whether holes were drilled in the basin area or on the bordering terraces.

The altitude approaches 5,000 feet; as a result, the climate is characterized by comparatively short, mild summers and fairly severe winters. Summer temperatures seldom exceed 90°, whereas winter temperatures may reach minus 30° for short periods. Thunder showers are not uncommon during the summer, and heavy rainstorms can be expected in the early spring and late fall.

#### PROPERTY AND OWNERSHIP

The areal extent of the property investigated comprised about 2,000 acres. Except for the swampy area in the river bottom, the acreage is used for agricultural purposes.

The owners include George M. and Floyd Loomis, G. E. and G. W. Davis, and Laurel Hansen. These persons may be contacted on their respective properties, or information as to where they live can be obtained at the farmhouses.

#### WORK BY THE BUREAU OF MINES

#### Drilling

The Bureau originally began drilling on the Gold Fork deposit June 24, 1951, under the Western Radioactive Minerals Exploration Program. These holes are designated in the tables as the G series. This phase of the drilling was done in the eastern portion of the deposit and was recessed on July 21, 1951, because most of the area was flooded as a result of irrigation. The second drilling phase, designated in the tables as the DH series, was done by the Bureau of Reclamation during November 1951 on the western extension of the deposit. Under a cooperative agreement, the Bureau of Mines supervised this drilling, processed the samples, and furnished the Bureau of Reclamation with analytical results. The purpose of drilling by the Bureau of Reclamation was to determine the amount of black-sand minerals that probably would not be mined as a result of flooding by the backwaters of the Cascade Reservoir. After 1951 the undrilled area, between the extreme eastern and western portions of the deposit, was about 1-1/2 miles in length (fig. 2); it was in this section that the Bureau of Mines began drilling August 8, 1956. Work was discontinued because of inclement weather on October 25. Holes drilled in this project are designated in the tables as the GF series.

Table 1 lists the holes drilled, depth in feet, percentage of core recovery, and number and weight of samples recovered from each.

Drilling in 1951-11/ by the Bureau of Mines and Bureau of Reclamation was done with a standard churn drill, using heavy-duty casing with an inside diameter of 6 inches. One 5-foot length of casing was equipped with a drive shoe having an inside beveled cutting edge of 7-1/2 inches. The depth of 1 hole was 140 feet. Ordinarily, the drilling was discontinued at 115 feet or less, except when bedrock was encountered at a shallower depth.

<sup>11/</sup> Kline, Mitchell H., Evaluation of Monazite Placer Deposits: Technical Services Inf. Service, U. S. Dept. of Commerce, RMO-908, April 1952, 16 pp.

TABLE 1. - Drilling data

				<u>'</u>
	-			Total dry weight
Drill-hole	Depth,	Core recovery,	Number of samples	of material
number	feet	percent	recovered	recovered, 1b.
$G-1\frac{1}{2}$	40	85.1	8	942
G-2	60	86.0	12	1,433
G-3	140	81.1	28	3,381
G-4	90	92.0	18	2,296
G-5	65	85.8	13	1,673
G-6	100	81.6	20	2,449
G-7	80	90.5	16	2,221
G-8	61	77.2	13	1,446
G-9	70	78.0	14	1,682
G-10	16	55.4	4	266
G-11	60	76.8	12	1,414
G-12	50	60.7	10	911
G-13	43	66.5	9	857
Subtotal	<del>875</del>	90.5	187	$\frac{057}{20,971}$
Arithmetical average	67.3	78.2	. 137	20,771
millimetical average	07,.5	70,2		
$_{GF-1}\underline{2}/\ldots$	40	66.2	8	609
GF -2	45	62.5	9	647
GF -3	45	64.1	9	663
GF -4	40	61.6	8	508
GF -5	35	69.1	7	556
GF-6	35	83.2	7	- 670
GF -7	35	68.3	8	550
GF -8	35	59.1	7	476
GF -9	40	75.2	8	692
GF-10	40	60.0	8	554
GF-11	35	69.6	7	552
Subtotal	425	97.0	85	6,477
Arithmetical average	38.6	67.2	. 05	0,4//
_	30.0	07.2	•	
DH-1 <sup>3</sup> /	70	100.8	14	3,474
DH-2	120	75.6	24	2,721
DH-3	110	59 <b>.</b> 5	22	, -
	110	56.3	22	1,963
DH-4	110	59.0	22	1,857
DH-6	110 115	70.7	22 23	1,946
			l .	2,439
DH-7	115 750	68.8	23 150	2,328
Subtotal		70 1	150	16,728
Arithmetical average	107.1	70.1		
Omend tetel	2 050		433	. ,, ,,,,
Grand total	2,030		422	44,176

<sup>1/</sup> G series of holes drilled by Bureau of Mines in 1951 for U. S. Atomic Energy Commission.

 $<sup>\</sup>frac{2}{3}$ / GF series of holes drilled by Bureau of Mines in 1956.  $\overline{3}$ / DH series of holes drilled by Bureau of Reclamation in 1951.

Drilling in 1956 was done with a smaller unit equipped with 5-inch casing, with 1 section having a 6-inch cutting shoe. Difficulty in pulling casing below a depth of 45 feet made it impracticable to drill deeper, although both upstream and downstream the previous drilling indicated that the black-sand values continued to a much greater depth (table 2).

Casing was driven 2-1/2 feet, and the resultant core rise was drilled and loosened to permit removal with a standard placer sand pump. Two 2-1/2-foot sections were combined into one 5-foot sample. Samples were dried in sheet-iron pans over an open fire and screened on a 1/4-inch-mesh screen. The oversize was weighed, logged, and discarded. The undersize was weighed, logged, tagged, and transported to the Spokane field laboratory for processing.

No attempt was made to recover the slimes from the drill holes. Previous drilling in other placer deposits indicated that only a trace to 1 percent of extremely fine black sands occurred in the slimes. Subsequent sampling on the dredges also indicated that the larger percentage of black-sand looses was in the smallest mesh sizes. For these reasons, it was not practical to take into account the small black-sand losses in the slimes to evaluate a placer deposit.

#### Laboratory Procedure at Spokane, Wash.

Samples recieved at the Spokane laboratory were reweighed and screened to plusand minus-16-mesh. All the oversize sample material was combined for each drill
hole and jig-concentrated to determine if there were any black sands that exceeded
16-mesh. In the case of the Gold Fork deposit, none were found in sufficient quantities to be of economic importance. Undersized material was rough-concentrated on
a No. 12 Wilfley table and reconcentrated on a small No. 13 Wilfley table to recover
a clean black-sand concentrate. This material was dried, weighed, and logged. The
sample was thoroughly mixed, and a 10-gram cut was taken. Magnetite was removed
with an Alnico magnet and the amount weighed and recorded. A small quantity of the
remainder was examined under a 40-power binocular microscope, an estimate of the
more common black-sand minerals was made, and the amounts of each were recorded.
Each sample was checked further with an ultraviolet light for zircon and scheelite,
also with a Geiger counter to determine the monazite equivalent 12/ in percent as
compared with the radioactivity of a standard monazite sample.

The dry weight of the alluvium taken from the drill holes was taken at 2,700 pounds per cubic yard, as indicated by tests made during the drilling program in 1951. Knowing the weight of the original gravel sample and the weight of the blacksand concentrate, the amount of black sand per cubic yard was determined as follows:

 $\frac{\text{Dry wt. of concentrate}}{\text{Dry wt. of sample}} \times 2,700 = \text{dry wt. of black sand per cu. yd.}$ 

(Weight of black sand per cu. yd.) x (estimated percent of any black-sand mineral) = 1b. of mineral per cy. yd.

The results of this laboratory examination are shown in table 2.

Monazite equivalent is a measure of the total radioactivity of a sample expressed as monazite. It is determined by comparing the radioactivity of an unknown sample with that of a monazite standard containing 4.2 percent  ${\rm ThO}_2$  and 0.13 percent  ${\rm U}_3{\rm O}_8$ .

TABLE 2. - Summary of laboratory estimates Weighted average for each drill hole

Drill	Total	<del></del>	Total	Field	estimate o	of mine	ral cont	tent in b	lack	<del></del>
hole	depth,	Calculated	black sand	sar	d, calcula	ated in	lb. per	cu. yd.		
No.	ft.	depth, ft.	1b. per cu. yd.	Ilmenite	Magnetite	Garnet	Zircon	Monazite	M.E. <u>⊥</u> /	Remarks
G-1	40	0-40	58.53	26.20	20.25	2.86	0.68	0.67	_	95 Au colors, 5-40 ft.
G-2	60	0-45 0-60	21.05 19.25	5.00 4.58	9.92 9.25	1.81 1.37	.50 .46	.27		8 Au colors, 35-40 ft.
G-3	140	0-40 0-85 0-140	24.79 29.02 20.33	6.67 14.22 10.13	11.16 6.98 1.44	2.91 3.40 2.43	.43 .36 .27	.47 .61 .46		2 Au colors. 122 Au colors, 55-90 ft. 138 Au colors.
G-4	90	0-35 0-90	14.11 27.40	4.84 13.01	5.50 6.15	1.58 5.14	.27 .20	.25		48 Au colors, 40-65 ft.
G-5	65	0-65	24.53	9.44	5.54	3.46	.26	.68	ì ·	
G-6	100	0-100	31.36	15.91	12,85	4.06	.08	.81		211 Au colors, 45-100 ft.
G-7	60	0-15 0-80	27.46 17.78	10.21 9.71	7.51 1.97	5.55 3.34	.27 .10	.46 .44		31 Au colors, 5-80 ft.
G-8	61	0-40 0-60	23.65 25.48	9.81 8.52	3.14 8.31	4.28 3.37	.19 .14	.63 .57		69 Au colors, 0-60 ft.
G-9	. 70	0-70	28.53	12.07	6.47	6.56	.25	.59		61 Au colors, 20-70 ft.
G-10	16	0-13	9.87	5.87	.41	2.29	.12	.28	}	4 Au colors, 0-13 ft.
G-11	60	0-60	48.83	18.04	17.24	5.27	.18	.60		104 Au colors, 30-60 ft.
G-12	50	0-50	7.74	1.92	2.27	1.31	.11	.16	ļ	5 Au colors, 0-10 ft.
G-13	43	0-15 0-43	26.46 18.04	12.25 5.29	.82 .46	11.54 4.62	.43 .36			
GF-1	40	0-40	6.81	3.61	.80	.92	2/Tr.	Tr.	0.25	Trace gold.
GF-2	45	0-45	23.00	13.80	Tr.	3.34	.46	.46	.75	Do.
GF-3	.45	0-45	7.43	4.07	.22	2.00	.15	.15	.25	
GF-4	40	0-40	10.48	5.03	.73	.84	.10	.05	.42	Trace gold.
GF <b>-</b> 5	35	0-35	19.92	7.98	Tr.	4.18	.20	.40	.69	
GF-6	35	0-35	21.12	6.54	11.40	1.06	.21	.21	.41	
GF-7	35	0-35	10.80	4.32	1.41	1.83	.22	.22	.35	
GF-8	35	0-35	7.93	1.37	2.64	.65	.55	.04	.22	·
GF-9	40	0-40	17.93	3.77	8.97	2.12	.18	.36	.47	
GF-10	40	0-40	28.83	13.84	6.92	3.55	.29	.29	1.12	<i>'</i>
GF-11	. 35	0-35	18.53	7.41	6.36	2.91	.18	.18	.49	·
DH-1	70	0-45 0-70	26.99 20.22	17.62 12.92	3.39 2.21	3.62 2.56	.19 .18			
DH-2	120	0-40 0-120	24.21 19.44	15.53 13.02	3.21 1.84	2.55	.29	.37		
DH-3	110	0-40 0-110	19.24 22.90	13.02 11.32	.96 5.04	3.24	.10	.49	,	
DH-4	110	0-45 0-110	15.96 10.87	10.29 7.02	1.62 .69	2.02 1.42	.31			
DH-5	110	0-40 0-110	13.41 12.70	8.37 7.79	.08 .06	1.76 1.83	.15	.27		
DH-6	115	0-10 -0-115	16.65 11.10	5.32 5.73	7.77 .92	4.42 1.60	.27			
DH-7	115	0-115	7.09	4.08	.24	.92	.34	.07		

Monazite equivalent. Trace.

After the preceding calculations had been made, the concentrates from each drill hole were composited and split on a Jones riffle; a representative sample was sent to the Bureau laboratory at Albany, Oreg. for petrographic, quantitative X-ray spectrographic, radiometric, and chemical analyses.

A diagrammatic flowsheet showing the procedure used for treating the alluvial samples in the Spokane laboratory is shown in figure 3.

#### Laboratory Procedure at Albany, Oreg.

Black-sand sample concentrates, upon arrival at the Albany laboratory, were split into two portions. One split was sent to the physics laboratory for X-ray spectrographic determination of the  $Cb_2O_5$  and  $Ta_2O_5$  content. (The lower limits of analysis by this method are 0.2 percent columbium and <0.5 percent tantalum.) The second split was sent to the petrographic section for a radiometric analysis to find the uranium equivalent  $(eU_3O_8), \frac{13}{2}$  followed by a petrographic estimation to determine the percentage of the various black-sand minerals present. The sample was then magnetically fractionized, using a Franz Isodynamic Separator, which produced four products: (1) Highly magnetic fraction; (2) titaniferous fraction; (3) ilmenite; and (4) nonmagnetic fraction.

The titaniferous fraction, if any, was ground and chemically analyzed for  ${\rm Ti0}_2$ . The ilmenite fraction was ground and split into 2 samples: 1 portion was returned to the physics section to be analyzed by the X-ray spectrographic method for percent  ${\rm Cb}_2{\rm O}_5$  and  ${\rm Ta}_2{\rm O}_5$ ; the remainder was sent to the chemical laboratory for determination of percent  ${\rm Ti0}_2$ , total Fe (occurring as both FeO and Fe2O3), CaO, MgO, MnO, Al2O3, and SiO2.

#### DESCRIPTION OF DEPOSIT

#### Size of Deposit

The deposit is exceptionally large, and the total potential mineral-bearing area was not delineated by drilling done by the Bureau of Mines and Bureau of Reclamation. The overall length is about 6 miles, and the width ranges from 1,000 to 4,000 feet where tested. The drill holes ranged from 16 to 140 feet in depth; bedrock was encountered in only 3 holes of the 31 drilled - drill holes G-6, G-8, and G-10 (fig. 2).

#### General Geology

No comprehensive geological report has been published on the placers of the Long Valley area, except for a classified report written in 1952 by Dr. J. Hoover Mackin of the Federal Geological Survey, on behalf of the Raw Materials Division of the United States Atomic Energy Commission.

The Gold Fork placer deposit, in the northern part of Long Valley, is on the western edge of the Idaho batholith that comprises an area of several thousand square miles in central and south central Idaho. Long Valley was formed as a result of a series of block faulting that has a north-south trend. The trough or basin formed has been filled with alluvium deposited by the North Fork of the Payette

<sup>13/</sup> Uranium equivalent is the measure of the total radioactivity of a sample expressed in percent U<sub>3</sub>O<sub>8</sub>; it includes radioactivity due to the presence of either uranium or thorium minerals or a combination of both.

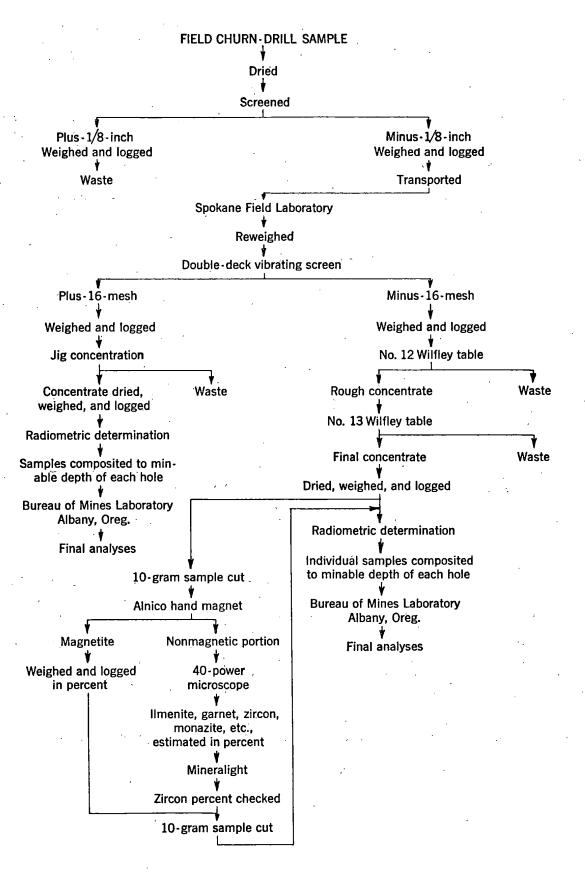


Figure 3. - Diagrammatic flowsheet showing treatment of alluvial samples.

River and its tributary streams, including the Gold Fork River, which has cut through the granitic type hills that bound the east side of the valley. West Mountain is a faulted block capped with basalt that rises abruptly several thousand feet above the valley floor on the west side.

On the north side of the deposit a granite ridge extends out into the valley for some distance; it has diverted the stream flow and resultant deposition of the alluvial gravels to the southward. The gravel in the upper horizon of the deposit was usually iron-stained and coarser than that at greater depths. This condition was particularly noticeable for about 1 mile downstream from the mouth of the gorge on the east side of the valley. At varying depths below 30 feet, clay was interbedded with fine sand and gravel. Quicksand and artesian water often were encountered beneath the clay beds.

The ilmenite and accessory black-sand minerals in the deposit were derived from the granitic type rock that was eroded within the Gold Fork drainage system. Gold colors were found in all the G series drill holes except G-13 (fig. 2).

Of the 31 holes drilled in the deposit, only G-1, G-8, and G-10 penetrated bedrock. One hole, G-3, was in gravel to a depth of 140 feet but did not reach bedrock. The amount of black-sand concentrates decreased noticeably below a depth of 105 feet.

The distribution of the black sands was erratic over the areal extent of the deposit and also within the vertical section of the individual drill holes. Drilling indicated that the heaviest black-sand concentration was on the south side of the present channel of the Gold Fork River.

#### Mineralogy

Table 3 shows the results of a screen analysis of composited black-sand concentrates from drill hole GF-9. The quantity of the more common black-sand minerals occurring in the various sizes from minus-14-mesh to minus-200-mesh was estimated in percent and recorded. Only 2.65 percent of the material was plus-28-mesh, and 7.87 was finer than 150-mesh. About 95 percent was in the minus-28-mesh - plus-150-mesh size range.

A mineralogical determination, recorded in percent of composited black-sand concentrates from four individual drill holes, is shown in table 4. Ilmenite was the predominant mineral, followed by garnet, magnetite, and sphene. Fine gold was found in the deposit in several upstream drill holes. There did not appear to be any particular relationship between the type and quantity of the minerals present with respect to location of the drill holes in the deposit area. Distribution of the black sands as to quantity or mineral type in the vertical section of the drill holes did not appear to concentrate at any particular horizon.

A few basalt pebbles and cobbles were noted during the course of drilling, but for the most part the gravels included mostly quartz, feldspars, and micas.

TABLE 3. - Black-sand screen analysis

Sample: 2 pounds table concentrate

				Mineral c	);						
Screen size, mesh	Table concentrate,	Ilmenite, lb.	Ilmenite	Magnetite	Garnet	Zircon		azite  M.E. <u>2</u> /	Quartz, feldspar, etc.	Ilmenite, percent	Ilmenite, cumulative percent
- 14 + 20	0.010	0.00250	25	12	40	0	0	<u>3</u> /Tr.	23	0.71	0.71
- 20 + 28	.034	.00680	20	25	40	. 0	1	2.7	14	1.94	2.65
<b>-</b> 28 + 35	.109	.02725	25	. 26	40	0	Tr.	1.6	9	7.78	10.43
- 35 + 48	.375	.05625	15	50	25	Tr.	0	1.1	10	16.07	26.50
- 48 + 65	.662	.09930	1.5	55	12	1	Tr.	1.5	17	28.37	54.87
- 65 + 100	.434	.08680	20	50	10	2	2	3.3	18	24.80	79.67
-100 + 150	.218	.04360	20	40	, 2	8	3	4.3	27	12.46	92.13
-150 + 200	.076	.01368	18	46	Tr.	15	3	5.1	18	3.91	96.04
-200	.077	.01386	18	51	Tr.	20	<sup>.</sup> 3	6.3	8	3.96	100.00

Visual monazite estimate. Monazite equivalent.

Trace.

TABLE 4	Mineralogical	composition,	visually	estimated	for composited
	blacl	k sand fracti	on of dri	11-hole sar	nples

	G-13	GF-3	GF-10	GF-11
	mineral,	mineral,	mineral,	mineral,
Species	percent	percent	percent	percent
Ilmenite	45.7	55	48	40
Garnet	27.4	21	18	- 15
Magnetite	1.9	2	24	26
Sphene	5 <b>.</b> 7	_ , 7	.3	3
Quartz	8.0	$\frac{1}{2}$ /S.A.	S.A.	S.A.
Epidote	3.2	$\frac{2}{2}$ /Tr.	1	Tr.
Kyanite	1.4	-		_
Ferromagnesium	1.7	Tr.	Tr.	Tr.
Zircon	1.1	1	1	1
Monazite	2.0~	2	1	1
Xenotime	.3	-	Tr.	Tr.
Allanite	Tr.	Tr.	Tr.	Tr.
Rutile	· -	Tr.	Tr.	< .5
Topaz	_	Tr.	Tr.	Tr.
Columbite	-	Tr.	· 🕳	<b>-</b> ·
Feldspar	_	Tr.	Tr.	Tr.
Apatite		_	Tr.	Tr.
Radioactive opaques	- '	Tr.	Tr.	Tr.

<sup>1/</sup> Small amount.

#### ANALYTICAL RESULTS

Results after the black-sand samples were fractioned and the analyses made are listed in table 5. The ilmenite fraction in the black sands ranged from 1.82 to 13.84 pounds per cubic yard. The average titanium dioxide content was 45.3 percent; iron content averaged 34.9 percent. Minor amounts of CaO, MgO, MnO,  $Al_2O_3$ , and  $SiO_2$  were present in the ilmenite fraction, which may occur in part either as discrete minerals or as a constituent part of the ilmenite. Columbium and tantalum were found to occur in the ilmenite; 14/ but unless a method for recovering it is developed, it has neither economic nor strategic importance.

 $<sup>\</sup>overline{2}$ / Trace.

<sup>14/</sup> Banning, L. H., Hergert, W. F., and Holter, D. E., Electric Smelting of Ilmenite Concentrates from Valley County, Idaho: Bureau of Mines Rept. of Investigations 5170, 1955, 18 pp.

TABLE 5. - Drill-hole-sample data, Gold Fork placer area, Valley County, Idaho

Drill- hole		Black sand,			ack-sand Ilmenite fraction,										
number	ft.	1b./cu. yd.	Cb <sub>2</sub> 0 <sub>5</sub>	Ta <sub>2</sub> 0 <sub>5</sub>	eU308 <u>1</u> /	1b./cu. yd.	Cb <sub>2</sub> O <sub>5</sub>	Ta <sub>2</sub> 0 <sub>5</sub>	T102	Fe	Ca0	MgO	MnO	A1203	SiO <sub>2</sub>
GF-1	40	6.81	0.45	<0.5	< 0.01	3.61	0.40	<0.5	45.8	26.0	<0.10	0.04	1.41	0.50	0.16
GF-2	45	23.00	.40	< .2	.03	13.80	,30	< .2	45.8	37.0	< .10	<.10	3.51	.48	4.50
GF-3	45	7.43	.35	< .5	< .01	3.72	.40	< .5	46.5	26.2	< .10	.07	2.00	.60	.17
GF-4	40	10.48	.30	< .5	.03	5.03	.30	< .5	45.8	36.6	.39	.23	2.61	1.40	<.20
GF-5	35	19.92	1.10	< .2	.03	13.74	.40	< .2	46.5	36.5	< .10	<.10	3.41	1.54	4.14
GF -6	35 .	21.12	<.20	< .5	.02	6.55	.40	< .5	44.2	37.4	.29	<.10	3.78	.20	2.92
GF-7	35	10.80	<.20	< .5	.02	4.32	.33	< .5	45.1	36.7	.29	<.10	4.53	.16	4.26
GF-8	35	7.93	<.20	< .5	.02	1.82	<.20	< .5	44.9	35.6	-	-	-	-	-
GF-9	40	17.93	<.20	< .5	.02	3.77	.44	< .5	43.6	37.5	.22	<.10	3.62	.30	4.84
GF-10	40	28.83	<.30	< .5	.02	13.84	.43	< .5	45.1	37.0	.29	<.10	3.86	.26	6.26
GF-11	35	18.53	.60	< .2	<.02	7.41	.30	< .2	44.8	37.5	< .10	<.10	3.27	.44	3.12

1/ Uranium equivalent.