UNDERGROUND MINE DRAINAGE CONTROL SNOWY CREEK-LAUREL RUN, WEST VIRGINIA FEASIBILITY STUDY



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UNDERGROUND MINE DRAINAGE CONTROL SNOWY CREEK-LAUREL RUN, WEST VIRGINIA FEASIBILITY STUDY

bу

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Contract No. S-802644

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FOREWORD

When energy and material resources are extracted, processed, converted, and used, the related pollutional impacts on our environment and even on our health often require that new and increasingly more efficient pollution control methods be used. The Industrial Environmental Research Laboratory - Cincinnati (IERL-Ci) assists in developing and demonstrating new and improved methodologies that will meet these needs both efficiently and economically.

The eastern United States has significant acid mine drainage problems as a result of underground coal mining. A major portion of these mines are no longer active. Techniques for controlling this pollution are limited, because of technical problems and cost. This study was undertaken to evaluate the feasibility of several innovative abatement methods. The Snowy Creek-Laurel Run basin, West Virginia, was selected for the study. The results of the study were that a lake to control mine pool level, a continuous clay core dam, and marble wall bulkhead seals were best suited for this watershed.

The results of this study should be of interest to those persons planning abatement programs for abandoned underground mines and to 208 planning agencies. For further information contact the Resource Extraction and Handling Division.

David G. Stephan
Director
Industrial Environmental Research Laboratory
Cincinnati

ABSTRACT

A study was conducted at the Snowy Creek - Laurel Run basin near Terra Alta, West Virginia, to determine the feasibility of demonstrating mine drainage control by known abatement techniques in abandoned coal mine areas having shallow overburden.

The basin contains two abandoned mining complexes that have extensively deep-mined the Lower Kittanning coal found in the Mount Carmel syncline. Associated mine pool discharges are responsible for 90 percent of AMD pollution in Snowy Creek which discharges into the Youghiogheny River (now being considered as a part of the National Wild and Scenic Rivers System). Only one-third of the Snowy Creek - Laurel Run basin is affected by AMD.

Additional inundation and stabilization of the mine pools were judged necessary to reduce the AMD pollution. The recommended approach was to utilize continuous clay core dams, a mine pool level control lake and movable wall bulkhead seals to increase the size of the mine pools. It was felt that this abatement approach was feasible.

This report was submitted in fulfillment of Contract No. 802644 by Baker-Wibberley & Associates, Inc., Hagerstown, Maryland, for the West Virginia Department of Natural Resources under the sponsorship of the U.S. Environmental Protection Agency. This report covers the period April 1, 1975 to March 1, 1977.

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CONVERSION TABLE

Divide	by	To Obtain
(Metric Units)	(Conversion)	(English Units)
cubic meters	1.308	cubic yards
cubic meters/minute	1.700	cubic feet/second
degrees Celsius	$(C \times 1.8) +32*$	degrees Fahrenheit
hectares	.405	acres
kilograms	.4536	pounds
kilometers	1.609	miles
meters	.9144	yards
metric tons	.907	short tons
nillimeters	25.4	inches

* Actual Conversion, not a division

This report was prepared during the period of national conversion to the metric system. Wherever practical, metric units are used. A few concessions were necessary for ease of readability. For example, elevations are quoted in feet in the absence of suitable metric topographic controls or mapping. Also, published tabulations or records are maintained in their quoted form.

ACKNOWLEDGEMENTS

Baker-Wibberley & Associates, Inc. wishes to acknowledge all and to express our appreciation to all who provided assistance during the completion of this project. Special gratitude is extended to:

- The property owners in the Snowy Creek Laurel Run basin.
- U.S. Environmental Protection Agency; Robert B. Scott, Project Officer.
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- Grafton Coal Company, Clarksburg, West Virginia; Jim Roy.
- The project team, Richard G. Beegle, Project Manager; Michael M. Dreisbach, Project Coordinator; Richard C. Ely, Mining Engineer; and Edwin F. Koppe, Geologist.

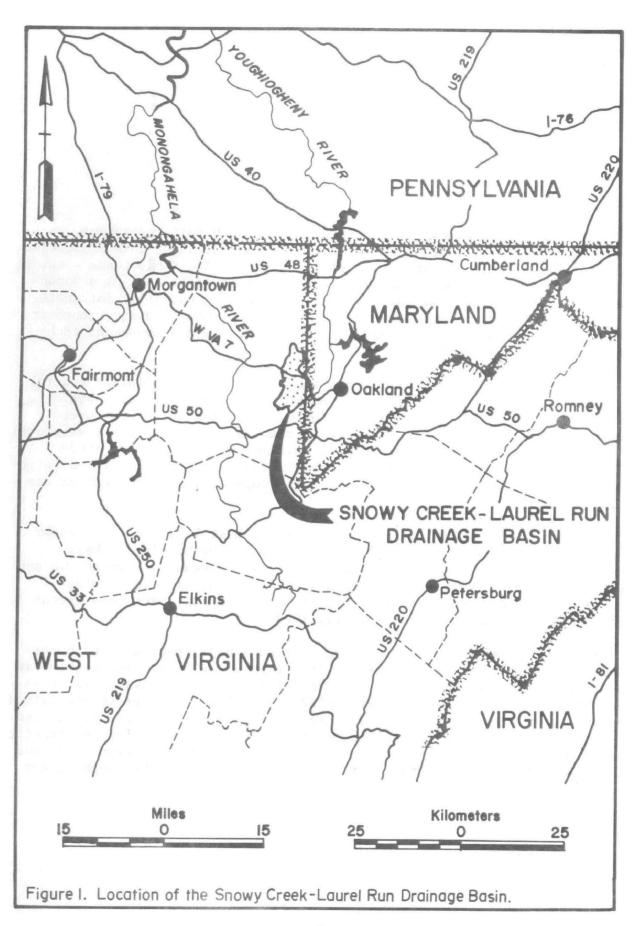
INTRODUCTION

The effects of abandoned mine drainage have been felt throughout Appalachia and West Virginia is no exception. Many areas of this State have been adversely affected by abandoned mine drainage. The Snowy Creek - Laurel Run basin was chosen as an excellent example that could be used as a demonstration project for determining methods with the best potential for abatement control designs. The demonstration of an abatement technique, once it has been proven effective, can serve as a model for future abatement technology under similar conditions.

This study has investigated in great detail existing pollution control laws affecting the basin, the physical environment (including the physiography and geology), present and past mining activities, water resources (including its climatology and hydrology), chemical analyses of the waters in the subject basin and the socio-economic environment found in the study basin. Investigations into these areas yielded a wealth of background information from which preliminary engineering of proposed abatement designs was developed.

The feasibility study is in Preston County, West Virginia. The Snowy Creek - Laurel Run basin (Figure 1) is situated on the Maryland - West Virginia border approximately 48.3 km (30 miles) southeast of Morgantown and south of the Pennsylvania - West Virginia border. The effects of mine drainage result from two major sections of the basin that have been extensively deep mined beginning in the early 1890's.

The Snowy Creek - Laurel Run basin is also a major headwater tributary draining into the Youghiogheny River. At present, the Upper Youghiogheny is being considered as a part of the National Wild and Scenic Rivers System as established by Congress in 1968 (PL 90-542). In this act Congress proclaimed that these scenic rivers, "shall be preserved in free-flowing condition, and that they and their immediate environments shall be protected for the benefit and enjoyment of present and future generations." Thus this basin study has a direct bearing on the "Scenic Rivers" classification for the Youghiogheny River.



CONCLUSIONS

- 1. This feasibility study proved that a demonstration project to abate AMD pollution in the Snowy Creek Laurel Run basin is feasible. Utilizing a continuous clay core dam and a mine pool level control lake will abate abandoned mine drainage in areas of mining having a shallow overburden.
- 2. Two major sources of acid mine drainage are found in the Snowy Creek Laurel Run basin-the Banner Mine and the Lima Mine. The Lima Mine pool is controlled by an outfall at an elevation of 743 m (2,438 ft). Acid mine drainage from the Lima Mine pollutes the portion of the study basin drained by Snowy Creek, above Laurel Run. The Banner Mine pool is controlled by the main borehole discharge located near the Maryland West Virginia border at an elevation of 729 m (2,393 ft). Discharges associated with the Banner Mine complex pollute the portion of the study basin drained by Laurel Run.
- 3. Additional inundation and stabilization of the Lima and Banner mine pools is necessary to reduce acid mine drainage pollution. Abatement at the Lima Mine is possible by constructing a subsurface dam (continuous clay core dam). At the Banner Mine, a mine pool level control lake will inundate additional deep mine workings and stabilize the mine pool. Implementation of these two projects will eliminate 90 percent of acid mine drainage discharges in the Snowy Creek Laurel Run basin.
- 4. Areas affected by past surface mining will remain above the proposed mine pool level control lake of the Banner Mine at an elevation of 750 m (2,460 ft) and will require regrading and revegetation to improve the aesthetic appearance of the basin.
- 5. Previous studies by the Maryland Water Resources Administration (WRA) indicate that the Youghiogheny River is severely degraded by acid mine drainage from the Snowy Creek Laurel Run Basin. Data collected during this study shows that Snowy Creek contributes a mean acid load of 4,663 kg/day (10,282 lb/day) to the Youghiogheny River.
- 6. The proposed abatement project will eliminate at least 3,437 kg/day (7,383 lb/day) of the acid load resulting from abandoned mine drainage in the Snowy Creek Laurel Run Basin. This reduction will reduce mean acid loadings on the Youghiogheny River by an equal amount. The affected section of the Youghiogheny River from its confluence with Snowy Creek

- to its confluence with the Little Youghiogheny River, some 7 km (4.4 miles) can now be improved.
- 7. Abatement of abandoned mine drainage will result in a cleaner water resource and a better environment for the Snowy Creek Laurel Run Basin and the Youghiogheny River. By cleaning up 6.3 km (3.9 miles) of stream in Maryland, 1.5 km (0.9 miles) of stream in West Virginia, and creating a new lake having 17.4 km (10.8 miles) of shoreline, improvements to fish and wildlife habitats will be expected.
- 8. The development of a wildlife refuge and/or recreational facilities may result as secondary impacts of the proposed project.
- 9. Definite social and economic gains will result from the cleaner environment made possible through the demonstration project. Protection of the Youghiogheny River and its environment will be achievable.

RECOMMENDATIONS

- Approval and funding should be granted to proceed with the detailed design engineering, construction, and monitoring phases of the demonstration project.
- 2. Reclamation of the Snowy Creek Laurel Run Basin should be accomplished according to the following priorities:
- a. Reclamation of the Snowy Creek portion of the basin should be accomplished by constructing a continuous clay core dam in the vicinity of the Lima Mine to eliminate acid mine drainage responsible for the pollution of the creek above Laurel Run.
- b. Reclamation of the Laurel Run portion of the basin should be accomplished by constructing an earth dam that will create a mine pool level control lake to raise the Banner Mine pool which is responsible for 90 percent of the acid mine drainage in Laurel Run.
- c. Abandoned surface mines in the Arnold Run and Freeport areas, should be regraded and revegetated above the impoundment shoreline.
- 3. A monitoring program should be constructed for sampling immediately above and below areas under abatement and at the mouth of Snowy Creek to assess the effectiveness of the project.
- 4. A new movable wall bulkhead mine seal design should be employed for seals of underground entries in the Ashby-Pendergast mine.
- 5. Interstate agreements on funding, acquisition, responsibilities and control should be set forth between the State of West Virginia and the State of Maryland as soon as possible to implement the demonstration project.
- 6. Involvement of Federal and State conservation agencies (such as U.S. Fish and Wildlife) and/or regional planning agencies will prove benefical toward the possible development of secondary benefits derived from the abatement project.

JURISDICTIONAL FRAMEWORK

This section establishes the legal authority for the State or for other agencies. A review of agencies involved with acquisition, design, construction and protection, and the relevant regulations under which they operate will be included to determine how effectively the demonstration project could be implemented under such agencies and standards.

COGNIZANT AUTHORITY

The following report has been conducted under the direction of the U.S. Environmental Protection Agency (EPA). The Water Pollution Control Act of 1972, PL 92-500, grants EPA its authority in water pollution matters. Section 107 of the Water Pollution Control Act, "Mine Water Pollution Control Demonstrations," addresses itself to selecting project watersheds to be used as examples of techniques developed to control mine drainage pollution. In the selection of project watersheds, EPA is directed to give "preference to areas with the greatest present or potential value for public use for recreation, fish and wildlife, water supply and other public uses."

The Snowy Creek - Laurel Run Demonstration Project is funded through a grant to the West Virginia Department of Natural Resources. The Department of Natural Resources is responsible for the supervision and administration of the demonstration project. Legislative authority to conduct such projects is granted to the Department of Natural Resources in the form of the Surface Mining and Reclamation Act and the Water Pollution Control Act of the Code of West Virginia. Administration of these laws under the Department of Natural Resources is performed by the Division of Reclamation and the Division of Water Resources.

The Division of Reclamation was established by the Surface Mining and Reclamation Act and is responsible for administering and enforcing all laws related to surface mining. The Division has control over land, water, soil restoration, and reclamation of all surface-mined lands. Section 20-6-3 of the Surface Mining and Reclamation Act grants the Division of Reclamation this authority. 1

The Division of Water Resources is charged with the Administration and enforcement of the Water Pollution Control Act, Article 5A, Chapter 20 of the Code of West Virginia, and it receives its authority from this Act. Sections 20-5A-2, 20-5A-3, 20-5A-3a and 20-5A-4 of the Water Pollution

Control Act set the parameters that guide the Division of Water Resources. The Division of Water Resources has been designated as the water pollution control agency for the State of West Virginia.

In summary, the Department of Natural Resources will be totally responsible for the implementation of the demonstration project, either through the Division of Water Resources or the Division of Reclamation or both.

EXISTING AND PROPOSED STANDARDS

Snowy Creek and Laurel Run are considered public streams, and they are therefore subject to the water quality standards regulated by the Division of Water Resources. The Administrative Regulations of the State of West Virginia, Series I, Sections 3 and 5 outline the general water quality standards applicable to the State waters. Section 3, "General Conditions Not Allowable in State Waters," establishes the general parameters on water quality standards for the State. Section 5, "Acid Mine Drainage Control Measures," sets forth specific conditions applicable to acid mine drainage discharges.

Snowy Creek, a headwater tributary of the Youghiogheny River, is also subject to more stringent water use and water quality criteria as established by Sections 6 and 13 of the West Virginia Administrative Regulations, Series II. Section 6, "General and Water Use Categories," defines the types of water uses in the State. Section 13, "Water Uses and Water Quality Criteria," develops the water quality standards that apply to all tributaries of the Youghiogheny River that are interstate with Maryland and Pennsylvania. Table 1 is a summary of water quality standards pertaining to acid mine drainage discharge as found in Section 13.2

SITE ACQUISITION

Abatement plans for the Lima Mine and Banner Mine areas will require that limited amounts of property be procured and that releases be obtained to perform certain segments of construction on lands not purchased.

Land acquisition is not a major concern in the project. Both the State of West Virginia and the State of Maryland have the legal capabilities to acquire land for such a project. A joint venture between Federal and State governments is considered necessary in order to fund the cost of the project.

Abatement at the Lima Mine will require that a release or permission to execute construction be obtained from the individual property owners. Certain segments of the Lima Mine abatement design may be executed by the active surface operator. Releases on approximately 4.9 ha (12 acres) will be required at the Lima Mine site. Acquisition of these releases should be obtained by the West Virginia Department of Natural Resources.

Constituent	Concentrations
Dissolved Oxygen	Not less than 5 mg/l at any time. Values normal for the waters in the area in question; however, generally held between 6.0 and 8.5, except streams carrying significant quantities of acid mine drainage shall have a pH of not less than 5.5.
Threshold Odor	Not to exceed a threshold odor number of 8 at 40°C. as a daily average.
Toxic Substance	Not to exceed 1/10 of the 96-hour median tolerance limit.
Nitrates	Not to exceed 45 mg/l. Not to exceed 100 mg/l. Not to exceed .001 mg/l. Not to exceed 1.0 mg/l. Not to exceed 1.0 mg/l. Not to exceed .01 mg/l. Not to exceed .01 mg/l. Not to exceed .50 mg/l. Not to exceed .01 mg/l. Not to exceed .05 mg/l.

Note: In special cases where the facts warrant, more stringent standards, or exceptions to the above standards, may be established in the individual case with the approval of the Environmental Protection Agency.

Source: Section 13 of the West Virginia Administrative Regulations, Series II, "Water Uses and Water Quality Criteria."

The Banner Mine abatement design will require approximately 270.3 ha (668 acres) to relocate an existing right-of-way and provide an impoundment area elevating the proposed mine pool. Areas located below an elevation of 749.8 m (2,460 ft) immediately behind the breast of the proposed earth dam, will be inundated. A deep mine seal will be required at the Ashby-Pendergast Mine; therefore a release to enter and construct such a seal will be necessary. Regrading and revegetation at the Arnold Run and Freeport areas will also require a release to enter and perform necessary work.

The Banner Mine abatement design project is situated on the Maryland-West Virginia border. The breast of the proposed dam is located in Maryland, and the impoundment is located largely in West Virginia. The Ashby-Pendergast Mine is located near Crellin, Maryland. Therefore, while the

impact of the abatement design and its physical construction are of interstate importance, acquisition of land and releases to perform work would require interstate cooperation between the State of Maryland and the State of West Virginia.

AUTHORITY FOR FUNDING

The abatement project for the Snowy Creek basin is considered an interstate project. The State of West Virginia, with the assistance of the U.S. EPA, has funded the feasibility study. Construction of the proposed abatement designs will greatly reduce acid mine drainage entering Snowy Creek and the Youghiogheny River, both of which are interstate streams. With the reduction of acid mine drainage originating in West Virginia, both the State of Maryland and of West Virginia derive benefits from the project.

Approximately 7.8 km (4.81 miles) of stream will be made cleaner with the implementation of the proposed abatement project. On Snowy Creek, 6.2 km (3.83 miles) of stream will be made cleaner of which 76 percent, or 4.7 km (2.92 miles) is found in Maryland. Laurel Run, below the breast of the proposed dam, will add another 1.6 km (0.98 miles) entirely in Maryland. In addition to these cleaner stream reaches shared by both Maryland and West Virginia, the proposed impoundment having 17.7 km (11.06 miles) of shoreline will also be a direct benefit to each state.

Finally with the reduction of acid mine drainage in Snowy Creek - Laurel Run, the Youghiogheny River can not return to its natural state, a goal Maryland is presently striving toward.

In view of the nature of the proposed program it is recommended that the State of West Virginia and the State of Maryland in cooperation with Federal agencies develop a joint method of funding the demonstration project.

WATER AND MINERAL RIGHTS

A mineral evaluation of the Snowy Creek Basin as well as the remainder of West Virginia is presently being conducted. Results of the evaluation will not be made public at this time. However, it has been determined that proposed abatement designs are in areas that have been economically mined out. Therefore, purchase of mineral rights has not been included.

The streams of Snowy Creek and Laurel Run are considered to be public streams, therefore private ownership is not an issue.

PREVENTION OF FUTURE POLLUTION

It has been determined that sufficient water quality standards exist and that proper authority has been granted the Division of Water Resources to administer and enforce said standards. Since the Division of Water Resources is the State's regulatory agency and has the authority to enforce water quality standards, implementation of the demonstration project in accordance to these regulations is anticipated.

Active mining in the project basin could be continued without any adverse effects to the proposed abatement project. Present standards and reclamation laws are sufficiently stringent to protect the proposed project.

The most recent mining permit issued to the mining operation within the basin included special conditions which were in harmony with the "clay core dam" abatement plan outlined in this project.

This addition of the continuous "clay core dam" concept, added by the operators in response to the project, and approved by the West Virginia Department of Natural Resources, exemplifies the concern and willingness to protect the project basin by all parties.

INVENTORY AND FORECAST

The section presents an inventory of physical, economic, and social conditions found in the Snowy Creek - Laurel Run Basin. An outline of physical conditions, including information obtained during subsurface explorations is presented. A complete water resource evaluation that emphasizes the results of the water quality network, which was monitored for 1 year, is summarized. Socio-economic conditions in the study basin and the effects of outlying areas on the basin are observed.

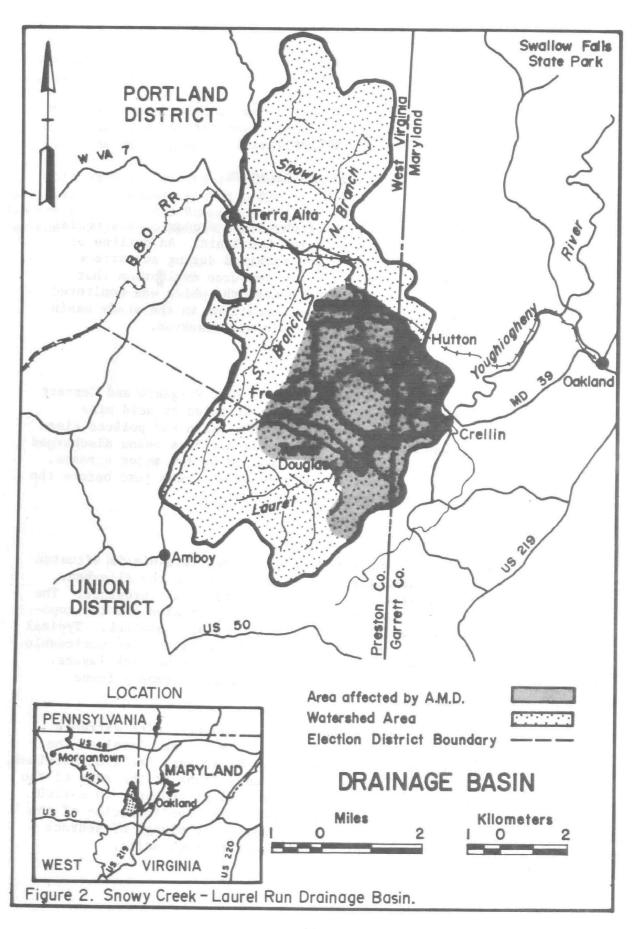
PHYSICAL CONDITIONS

The project is situated in Preston County, West Virginia and Garrett County, Maryland, as shown on Figure 2. Areas affected by acid mine drainage are situated in the lower portion of the basin and pollute clean water derived from the upper segments of the basin before being discharged into the Youghiogheny River. The basin is drained by two major streams, Snowy Creek and Laurel Run. Laurel Run enters Snowy Creek just before the Snowy Creek confluence with the Youghiogheny River.

Physiography

The Snowy Creek - Laurel Run study area in West Virginia is situated in the headwaters of the Youghiogheny River Basin within the Allegheny Mountain section of the Appalachian Plateaus physiographic province. The plateau in this area has been strongly dissected by erosion and the topography reflects the lay and alternation of hard and soft bedrock. Typical steep-sided, v-shaped valleys separated by rounded hilltops are noticeable only where the drainage cuts though resistant sandstone bedrock layers. Broader and more gentle rolling hills and valleys are commonly found elsewhere in the immediate survey area.

Overall topographic relief is about 250 m (820 ft), with extremes ranging from a low elevation of 727 m (2,385 ft) above sea level at the confluence of Snowy Creek and the Youghiogheny River near Crellin, Maryland, to a high elevation of 992 m (3,256 ft) on Brushy Knobs at the edge of the watershed to the southwest. In the immediate area of concern, 60 m (200 ft) to 120 m (400 ft) of relief is more common. The configuration of the watershed drainage system reflects both the northeast plunge of bedrock structure as well as basinward dips of resistant bedrock units.



Geology

The study area occupies the center and margins of the southwest-northeast trending Mount Carmel syncline. The syncline is bounded by the Briery Mountain anticline to the west and the Deer Park anticline to the east. There is a northeastward plunge to the syncline, as shown by structure contours plotted on the accompanying geologic map (Figure 3). The relatively broad and flat axis of the syncline here is bounded by a pronounced steepening of bed attitude toward the paralleling anticlines. About 260 m (853 ft) of coal-bearing strata are preserved in the syncline referable to the Pottsville, Allegheny, and Conemaugh groups of the Pennsylvania System (Figure 4).

The basal Pottsville Group is made up mainly of massive sandstone and siltstone, with only minor thin coal horizons. Coals in the Pottsville Group are unimportant; they tend to be less than 0.5 m (1.6 ft) and generally are not minable. Exposures were found at the Mercer and Quakertown horizons. No pollution problems were found.

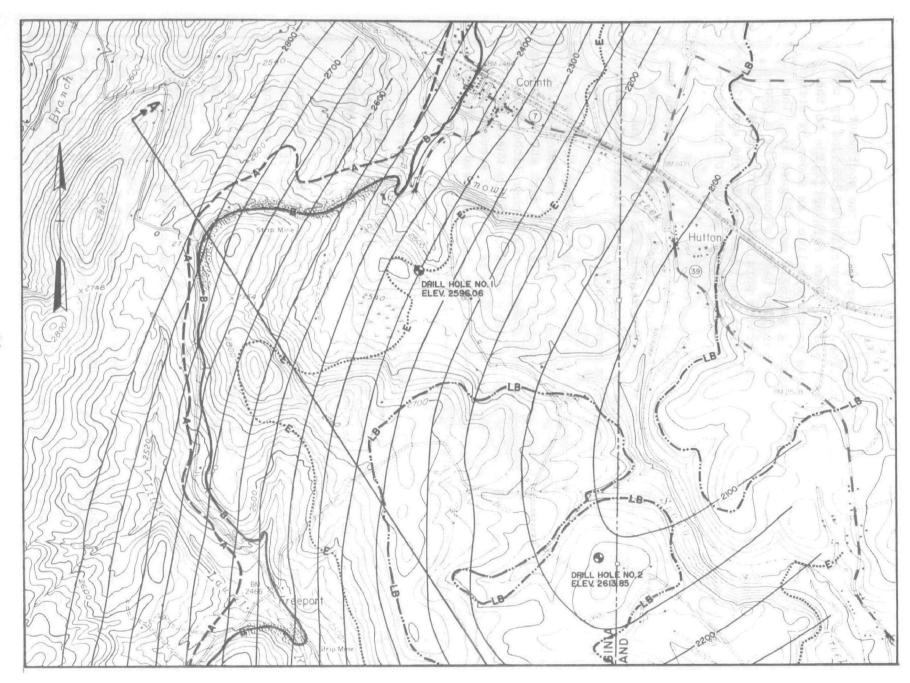
The Allegheny Group includes those strata found between the Lower Clarion and the Upper Freeport coal horizons. The group includes cyclic sequences of shale, sandstone, coal and clay. Light to dark shale, ranging from sandy shale to clay shale, is more abundant than other lithologies. Sandstone is common to the lower half of the Allegheny Group where it laterally interfingers with, or massively replaces, a shaly sequence. The persistent minable lower Kittanning coal horizon has been used as the key horizon, providing both stratigraphic and structural control in the study area.

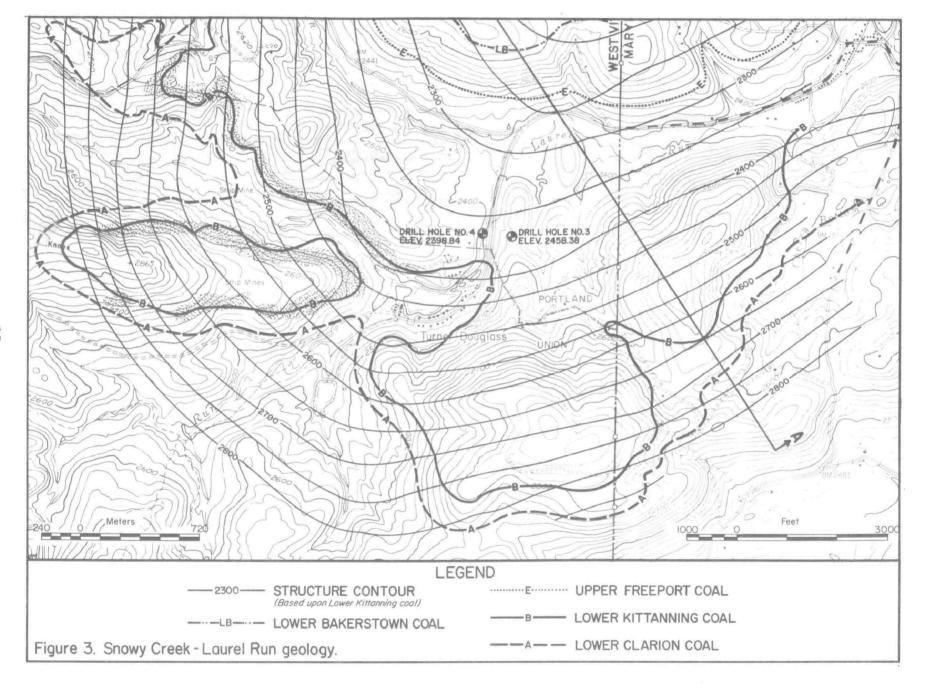
The Lower Clarion coal near the base of the group was reported as 0.3 to 1.2 m (1 to 4 ft) thick and of poor quality but has been prospected only at a few places. No significant mining of this bed was found anywhere in the study area. This coal is found 24 and 30 m (79 ft and 98 ft) below the Lower Kittanning coal datum.

The Lower Kittanning coal is the principal mined coal of the area. This coal is very persistent and generally consists of a split bed containing a middle 1.0- to 1.5- m (3 to 5 ft)- thick black shale parting. The total coal thickness, including partings, is about 3 m (10 ft). Generally, a massive to shaly sandstone 10 m (33 ft) or more thick is found within a few meters of the top of the coal.

Coals noted above the Lower Kittanning in the Allegheny group tend to be thin or lenticular and though prospected and mined very locally, they are of little importance to the project.

The Conemaugh Group in the study area includes all consolidated strata above the position of the Upper Freeport coal horizon. Originally 200 to 230 m (656 to 755 ft) thick, this unit has had all vestiges of its upper two thirds removed by erosion. This group is made up of shale, sandstone, and claystones with a few thin coals. Only the Lower Bakerstown





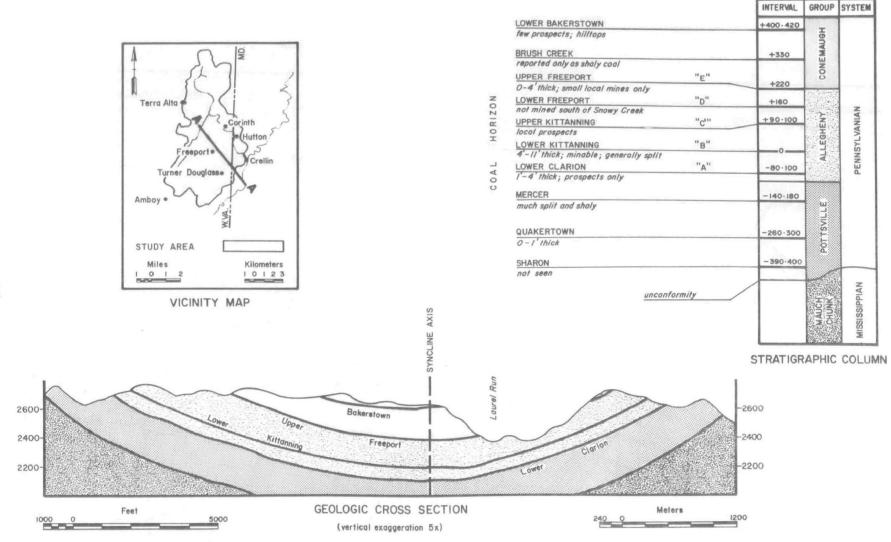


Figure 4. Geologic data for the Snowy Creek-Laurel Run basin.

coal has been prospected and opened for house coal. The Brush Creek horizon $35\ \mathrm{m}$ (115 ft) above the Upper Freeport coal, is usually a carbonaceous.

A thin alluvial fill is present in the Snowy Creek and Laurel Run valleys. The alluvium is seldom as much as 5 m (16 ft) thick in the center of either valley.

Although no true faults were found, a possible fault was reported in the mine workings and may have some displacement. The rocks of the area are fractured both by natural jointing and as a result of subsidence in some areas where the Lower Kittanning coal was extensively mined. Effects of subsidence were particularly noticeable in places where coal was withdrawn less than 30 m (98 ft) below surface. The opened fractures materially affect the hydrology in the vicinity of mining. In such areas, normal perched groundwater supplies are either meager or absent. Depending on the elevation, water either enters the mine or is discharged from the mine when fractures are present in drainage courses.

Subsurface Exploration

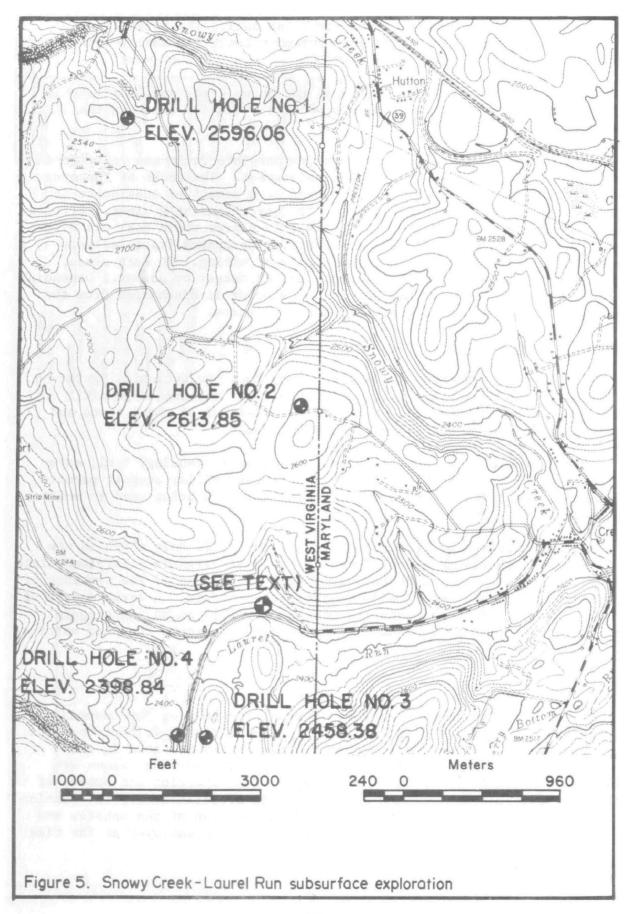
Test borings were made during the study to prove geological interpretations, to document the nature of the overburden, and to permit monitoring of the Banner Mine pool level. Locations of these tests and resultant logs are presented as Figures 5 through 9.

Broken strata were found above the B (Lower Kittanning) coal, especially in drill holes 3 and 4, where insufficient pillar support permits varying degrees of subsidence. Test Boring No. 3 is being used to monitor the level of the Banner Mine pool.

An attempt was made to rebore the existing drill hole which is now providing a discharge to Laurel Run near the West Virginia - Maryland State Line. The tools could not be held in alignment because of the presence of a concrete plug and lateral voids in the broken strata. A large amount of concrete had been injected into the hole during 1964 in an unsuccessful attempt to halt this flow. The flow had channeled around the plug material and widened the hole.

Mining

Extensive underground mining has taken place only on the Lower Kittanning coal, and all significant pollution is from this horizon. The extent of this mining and of peripheral strip mining is shown in the accompanying mine development map (Figure 10). The workings shown are principally taken from available mine maps. The direction and extent of some minor workings are assumed in the absence of satisfactory information (Figure 10). The assumed mains are based on position of the entries and reported extent or interpretations of mining methods employed at the time of development.



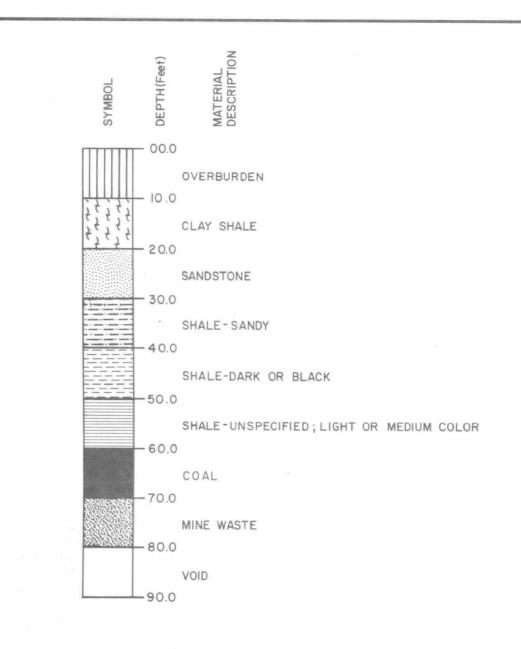


Figure 6. Legend for test boring logs.

BORING I

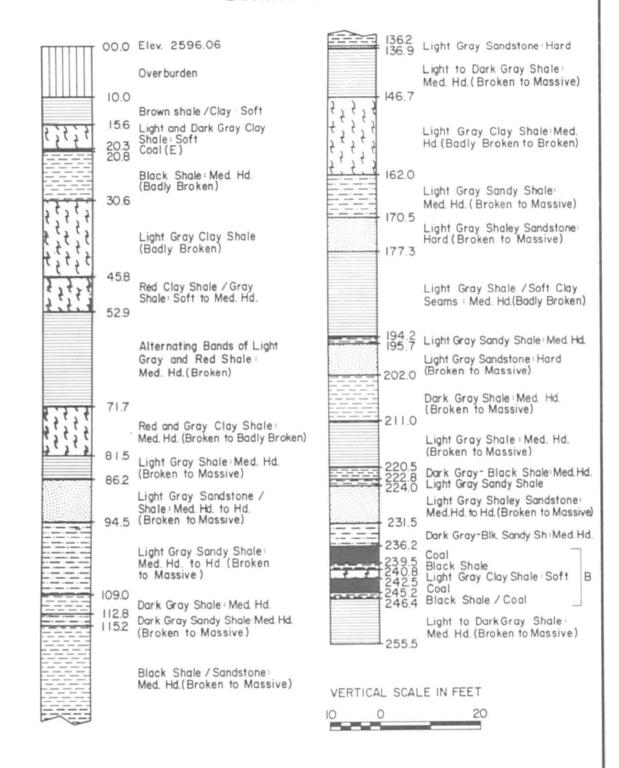
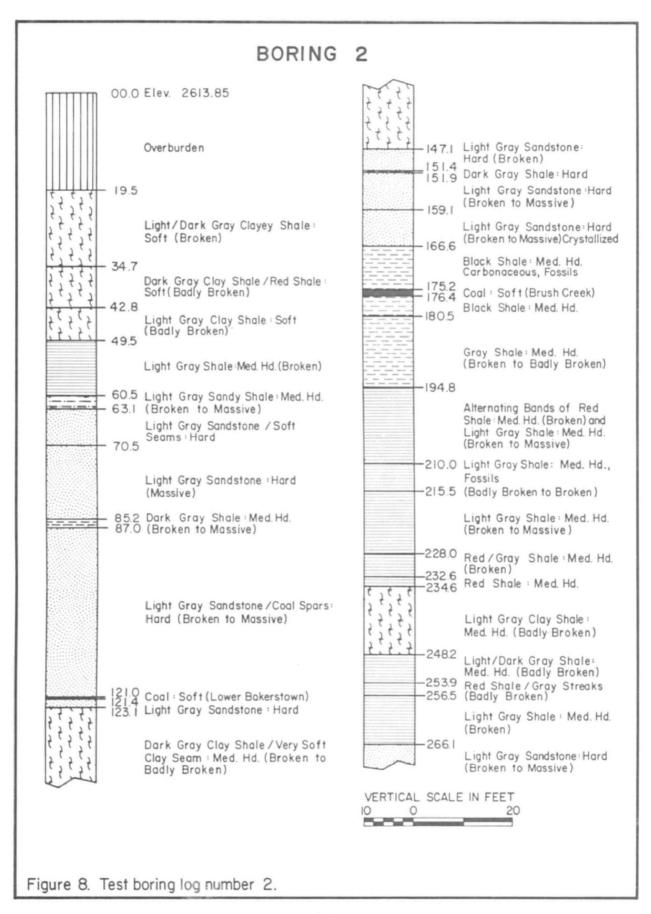
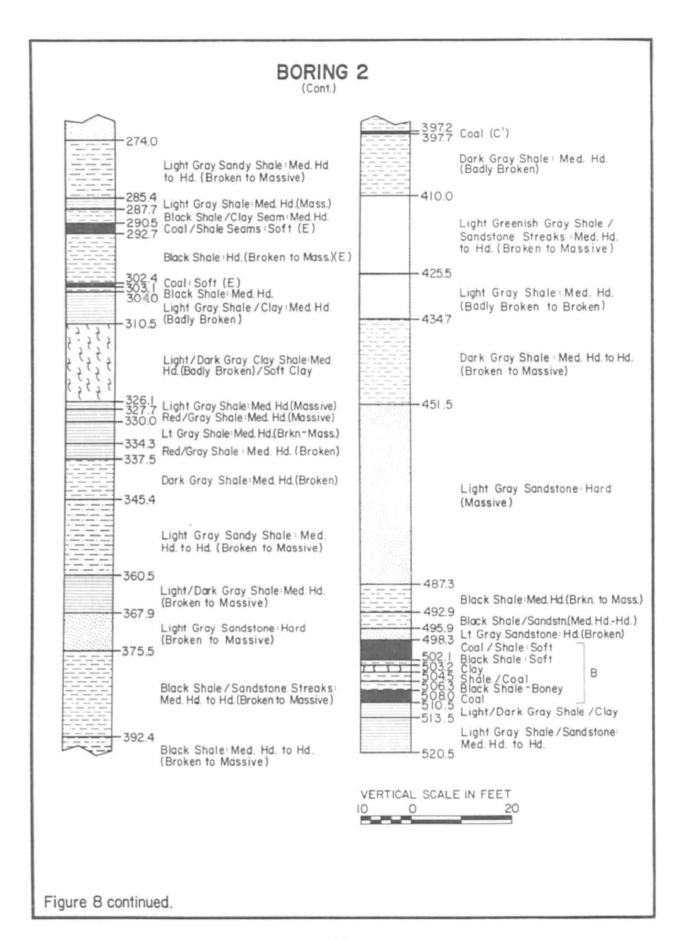


Figure 7. Test boring log number 1.





BORING 3

BORING 4

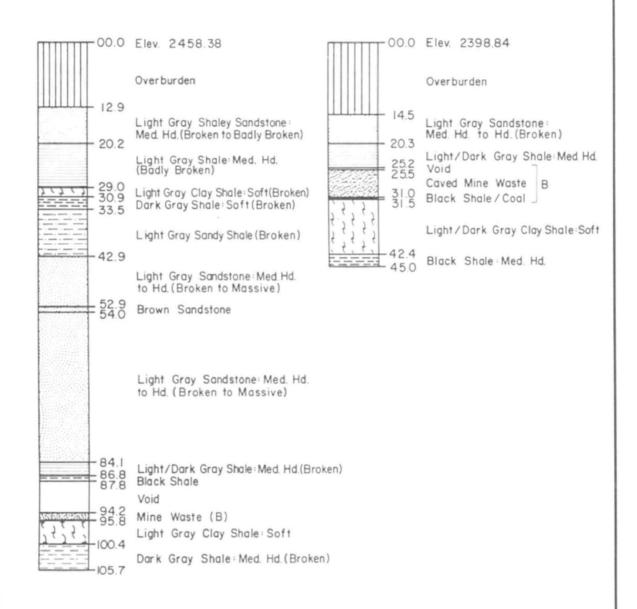
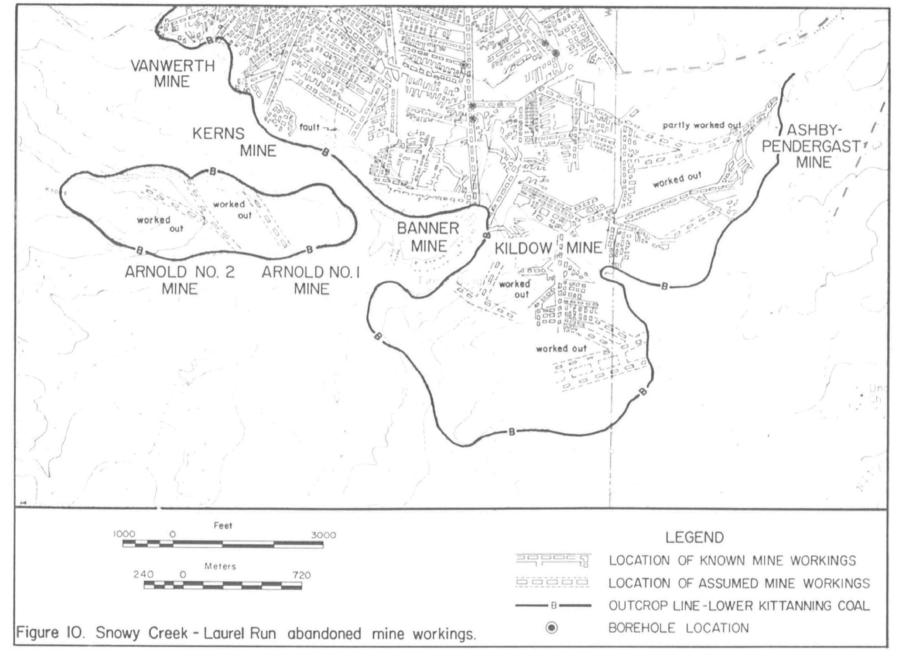




Figure 9. Test boring logs number 3 and number 4.



The presence of a mine pool in the Lima Mine about 14 m (46 ft) above the elevation of the Banner Mine pool strongly suggests that no connection exists between these pools at the present time.

Mining was started in the Snowy Creek - Laurel Run Basin at Corinth in 1897. The first mine, the Corinth Mine, was opened by the Oakland Coal and Coke Company in what they believed to be the Upper Freeport Coal seam (now identified as the Lower Kittanning Coal Seam).

The mine portal was located approximately 700 m (2,297 ft) southwest of the center of Corinth. The main heading was driven to the southwest to allow for natural drainage from the mine. Most of the mine development was to the west of the main heading, and mining was to the rise. Several butt headings were driven almost due south and extended far enough to come very close to if not actually intercepting, the workings in the Banner Mine.

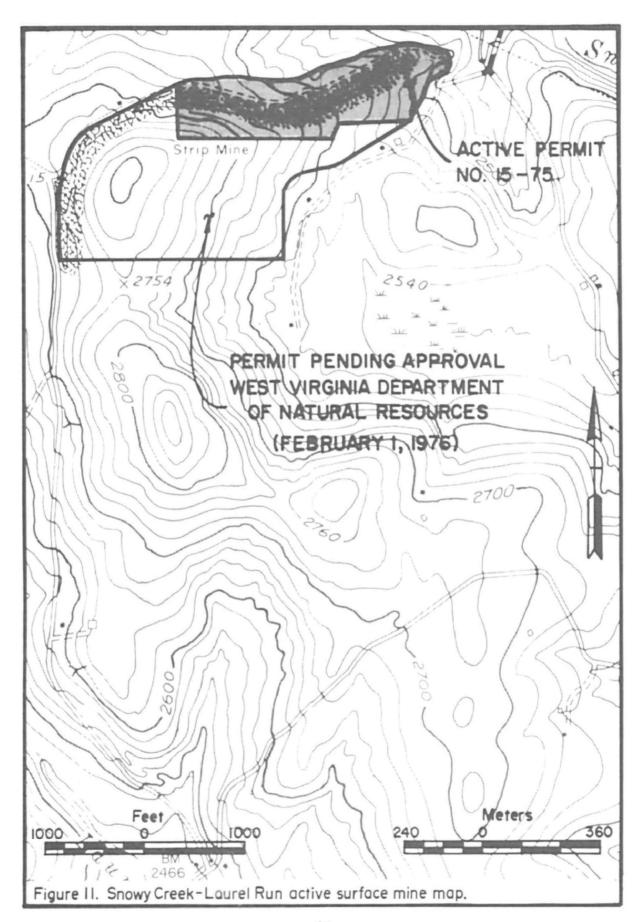
The Oakland Coal and Coke Company mined coal in the Corinth Mine until 1910, when the records show that mining was continued under the name of the Jorden Coal Company, which mined coal until 1912. From 1912 until 1922, the records show no production recorded from the Corinth Mine. In 1922, the Lindsey Coal Mining Company opened operations and named the mine the Lima No. 1. Production records continue from the Lindsey Coal Mining Company until 1932. Here again, no production records are available until 1942, when the Princess Pat Coal Company began reporting production and continued until 1947. This we believe was the last of the production from the deep mine.

In 1953, the L & L Coal Company and the Mersing Coal Company began a surface mining operation along the outcrop of the Lower Kittanning Coal Seam. Surface mining was carried on until 1960. In 1975, the Grafton Coal Company began surface strip mining operations very near the portal of the Lima No. 1 mine and continued the surface mine operation along the existing highwalls. All of these operations cut into the old mine workings of the Lima Mine. Figure 11 locates the present active surface mine operations of the Grafton Coal Company.

The majority of underground mining was done to the rise in this mine employing the room and pillar system. This was accomplished by driving the main heading into the body of coal on the rise of the seam (up dip). Branch entries (butt headings) were then driven off the main heading at approximately 107 m (350 ft) intervals. The butt headings were driven slighly up dip in order to provide natural drainage and to aid in hauling. The rooms were also driven from the butt headings on the rise (up dip) and were spaced at approximately 15 m (50 ft) intervals.

The area mined down dip is much smaller because of the inability to properly handle the water created by mining down structure. We feel that water rather than the quality of the coal was the controlling factor in the extent of mining to the east of the main headings.

By 1897, there were two commercial mines in production in the Turner Douglass area: the Arnold No. 1 and the Guthrie Mine (production figures



were not available until 1913). The original producer of coal was the Preston Lumber and Coal Company. In 1905, the Kendall Lumber Company purchased the Preston Lumber and Coal Company, and by 1919, coal producing was its main enterprise. At this time there were four active mines in the area—they were the Arnold No. 1 and No. 2, Guthrie, and Turner Douglass (later named the Banner Mine).

The main portal of the Banner mine is located at the east end of Turner Douglass and the main heading was driven almost due north and is over 3.2 km (2 miles) long. But headings were driven off the main heading to the left and right and rooms were then driven off the butt headings.

All of the work was done by hand in the Turner Douglass area until 1925, when the Kendall Lumber Company sold its holdings (which included the Turner Douglass Mine) to the newly formed Stanley Coal Company. The Stanley Coal Company, which named the Turner Douglass Mine the Banner No. 1, began a modernization program, and by 1927 it was 95 percent mechanized.³

As shown on Figure 10, the deep mine workings were extended almost to the "grass roots" on the western side of the main headings. It is also shown that the Nordic, Laurel Valley, Vanwerth, Kerns, Banner, Kildow, and Ashby-Pendergast Mines are all interconnected. We believe that the intent was at one time to cut through and connect the Lima Mine with the Banner Mine. Based on the mine development maps and productions reported by the Grafton Coal Company, these mines show about 75 percent recovery. The limiting factor in mining on the eastern side of the main heading was the amount of water that could be handled economically.

The Arnold No. 1 and No. 2 mines were not connected to the Banner Mine. (No mine maps were available for the area.) Mine production records indicate the area was partially mined out.

An area of Lower Kittanning Coal partially deep-mined is located in the study area south of Turner Douglass. The area has one small mine driven only a short distance and then abandoned. According to information furnished by local residents the thickness and quality of the coal changed in the area and was not mineable. This has not been verified by drilling and there does seem to be a potential mining area at this location.

Surface mining was extensively carried out in the Turner Douglass-Freeport area. Most outcrop coal was removed in surface mine operations. During surface mining many old workings were cut into, thus demonstrating the fact that deep mine headings were driven to the "grass roots."

Many of the mines were known by different names. Available mine production records for the Snowy Creek Basin are presented in Appendix A.

Soils

Soils in the study area are typically residual soils derived from interbedded acidic sandstone, siltstone, and shale. The Gilpin series dominates the uplands grading downslope to the related, poorly drained

Ernest and Atkins soils in the valley floor. The principal soils found are listed in Table 2.

The Gilpin soils are moderatly deep, well-drained silt loams with variable amounts of slabby or stony material present. This series locally includes patches of more sandy Dekalb soil on steeper slopes and patches of claypan Wharton soil in flat upland areas.

All of the soils are acidic, ranging from pH 4.2 to pH 5.0, and tend to be best suited for wildlife habitat, woodland, or pasture lands. Disturbed areas can be returned to usefulness with minimal fertilizing and liming.⁴

TABLE 2. SOILS CLASSIFICATION

Symbol Symbol	Name
Aa	Atkins silt loam
Bm	Brinkerton silt loam, 3-10 percent slopes
Bn	Brinkerton stony silt loam, 0-15 percent slopes
Ca	Calvin silt loam, 3-10 percent slopes
СЪ	Calvin silt loam, 10-20 percent slopes
Cd	Calvin silt loam, 20-30 percent slopes
Ce	Calvin silt loam, 20-30 percent slopes
Cf	Calvin silt loam, 30-40 percent slopes
Cg	Calvin silt loam, 30-40 percent slopes
Cm	Cavode silt loam, 3-10 percent slopes
Co	Cavode silt loam, 10-20 percent slopes
Ср	Clarksburg silt loam, reddish variant, 3-10 percent slopes
Db	Dekalb channery sandy loam, 10-20 percent slopes
Dc	Dekalb channery sandy loam, 20-30 percent slopes
Dg	Dekalb loam, 10-20 percent slopes
Dk	Dekalb loam, 20-30 percent slopes
Dr	Dekalb stony loam, 20-30 percent slopes
Du	Dekalb stony sandy loam, 5-20 percent slopes
Dv	Dekalb stony sandy loam, 20-30 percent slopes
Dw	Dekalb stony sandy loam, 30-40 percent slopes
Eb	Ernest silt loam, 3-10 percent slopes
Ef	Ernest stony silt loam, 3-20 percent slopes
Gk	Gilpin silt loam, 3-10 percent slopes
Gn	gilpin silt loam, 10-20 percent slopes
Go	Gilpin silt loam, 10-20 percent slopes
$G_{\mathbf{p}}$	Gilpin silt loam, 20-30 percent slopes
Gr	Gilpin silt loam, 20-30 percent slopes
Gv	Gilpin stony silt loam, 3-10 percent slopes
Gw	Gilpin stony silt loam, 10-20 percent slopes
Gx	Gilpin stony silt loam, 20-30 percent slopes
Gy	Gilpin stony silt loam, 30-40 percent slopes

(continued)

TABLE 2 (continued)

Symbol	Name		
Lickdale stony silt clay loam, 0-15 percent slopes			
Mb	Mine dumps		
Mc	Melvin silt loam		
Ra	Rayne silt loam, 3-10 percent slopes		
Se	Strip mine spoil		
Wa	Wharton silt loam, 3-10 percent slopes		
Wc	Wharton silt loam, 10-20 percent slopes		
Wd	Wharton silt loam, 10-20 percent slopes, severely eroded		
We	Wharton silt loam, 20-30 percent slopes		
Wf	Wharton silt loam, 20-30 percent slopes, severely eroded		

Source: United States Department of Agriculture, Soil Conversation Survey - Preston County, West Virginia

WATER RESOURCES

Climatology

The National Oceanic and Atmospheric Administration maintains a weather station at Terra Alta, West Virginia, in the northwest corner of the study basin. The exact location of the weather station has been moved once since 1952. Data collection, however, has remained in the same general location. Therefore the early gage known as Hopemont (from 1952 to 1964) and the present Terra Alta No. 1 gage were combined to obtain a longer historical data base. These climatological patterns are summarized by the normal values found on Table 3 and Figure 12. The normal values were then used as a base for comparison against the recorded values also expressed on Table 3 and Figure 12.

Temperature--

Generally the temperatures recorded during the sampling period were above normal. Average temperatures for February and March 1976 ranged far above normal. During this period, an extremely early spring thaw occurred, with temperatures reaching as high as 19° C (67° F) on February 29.

These warmer temperatures caused the existing accumulation of snow and ice to be melted, thus creating a high runoff on February 16. But because of the shortened period in which water would have been stored in the form of ice and snow, the runoff potential during this early spring thaw was held to a minimum.

In summary, the recorded temperatures were generally 2.3° C above normal. The early spring thaw was directly responsible for lower than anticipated flows recorded during the spring runoff.

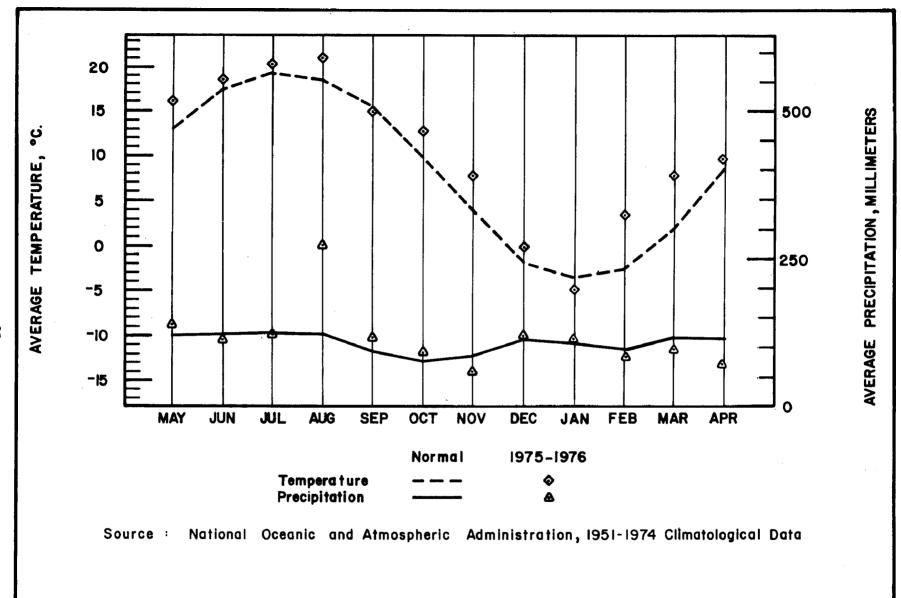


Figure 12. Climatological data for the Terra Alta, W.Va. weather station.

TABLE 3. CLIMATOLOGICAL DATA
TERRA ALTA, WEST VIRGINIA WEATHER STATION

	Temp	erature °C)		Rainfall (mm)	
Month	R	N	R	N	*
Mars 107E	16.3	12.2	136	120	40
May 1975		13.2			
June 1975	18.7	17.5	110	121	48
July 1975	20.2	19.3	121	122	36
August 1975	20.9	18.5	273	122	8
September 1975	14.9	15.5	116	94	28
October 1975	12.7	9.7	89	76	23
November 1975	7.8	3.9	55	85	85
December 1975	0.0	-2.0	120	109	42
January 1976	-3.9	-3.6	118	102	31
February 1976	3.1	-2.7	88	94	50
March 1976	6.8	1.9	97	117	62
April 1976	9.4	8.4	74	115	77
Average Total	10.6	8.3	1,397	1,272	

R = Value recorded during sampling program

Source: National Oceanic and Atmospheric Administration Climatological Data Summary.

Precipitation--

Precipitation values recorded were also generally above normal. The wettest month of this sampling period was August 1975, during which time 272.5 mm (10.73 in) of rainfall was recorded. This recorded amount was the second wettest August on record for the gage and was also the wettest location in the State during August 1975. Total precipitation recorded at the Terra Alta gage for the water year ending December 1975 was 1,599 mm (62.95 in).

Tabulation of total precipitation during the sampling period indicates that 1,397 mm (55 in) was recorded from May 1975 to April 1976. Precipitation data collected during the periods of August 11 to September 2, 1975, and December 26, 1975, to January 1, 1976, deserve special emphasis.

Beginning on August 11 and ending on August 17, 127.3 mm (5.01 in) of rainfall was recorded with 43.2 mm (1.7 in) recorded on August 14. This initial storm quickly saturated the dry soil and recharged the groundwater table. Four days later high stream flows were recorded, but they were less than anticipated because of the increased groundwater storage.

A second period of rainfall began on August 23 and ended on September 2. During this period, 132.3 mm (5.21 in) was recorded, with 45.7 mm (1.8 in) of it recorded on August 24. By this time, the groundwater table had

N = Normal value

^{* =} Percent of time rainfall can be expected to be exceeded.

been recharged and soil conditions were in a semisaturated state. Relatively high flows were therefore recorded on September 2.

The second period of marked precipitation occurred during the winter holiday season beginning December 26, 1975, and ending January 1, 1976. During this period, 90.4 mm (3.56 in) of rainfall was recorded, with 37.6 mm (1.48 in) of it occurring on January 1. This heavy rain, coupled with temperatures reaching 8° C (47° F) caused rapid melting of existing snows on the ground. Thus the highest flows for the entire study period were recorded on January 1.

In summary, historical data show (Figure 13) that the 1,397 mm (55 in) of precipitation recieved during the sampling period can be expected to be exceeded only 32 percent of the time.

Sampling Program

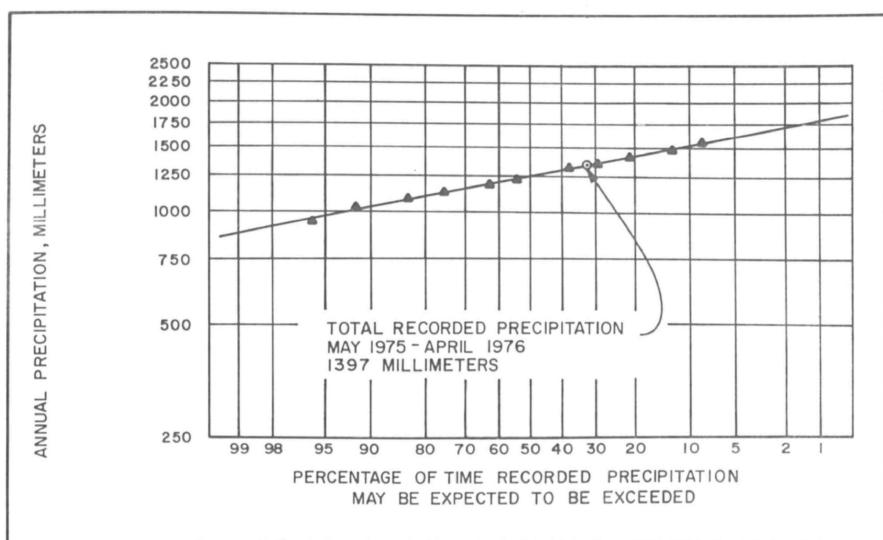
A detailed stream sampling network was constructed and maintained for 1 hydrologic year. Figure 14 identifies sample stations employed during the study. Appendix B defines each sampling station using the Universal Transverse Mercator system.

Beginning in May 1975 and ending in May 1976, over 900 sample locations were checked and over 850 chemical analyses were performed. Generally, routine chemical analysis for the basic mine drainage parameters of pH, acidity, alkalinity, sulfates, total iron, ferrous iron, specific conductance, aluminum, and manganese were conducted.

On each major pollution source, additional comprehensive water quality analyses were also conducted. These samples include the routine analyses mentioned above plus analyses for cadmium, calcium, chromium (hexavalant), chromium (total), copper, cyanide, lead, magnesium, mercury, potassium, zinc, arsenic, turbidity, total solids, suspended solids, total organic carbon, chloride, fluoride, chemical oxygen demand and hardness. Methods use in testing for these chemical parameters are a part of Appendix C.

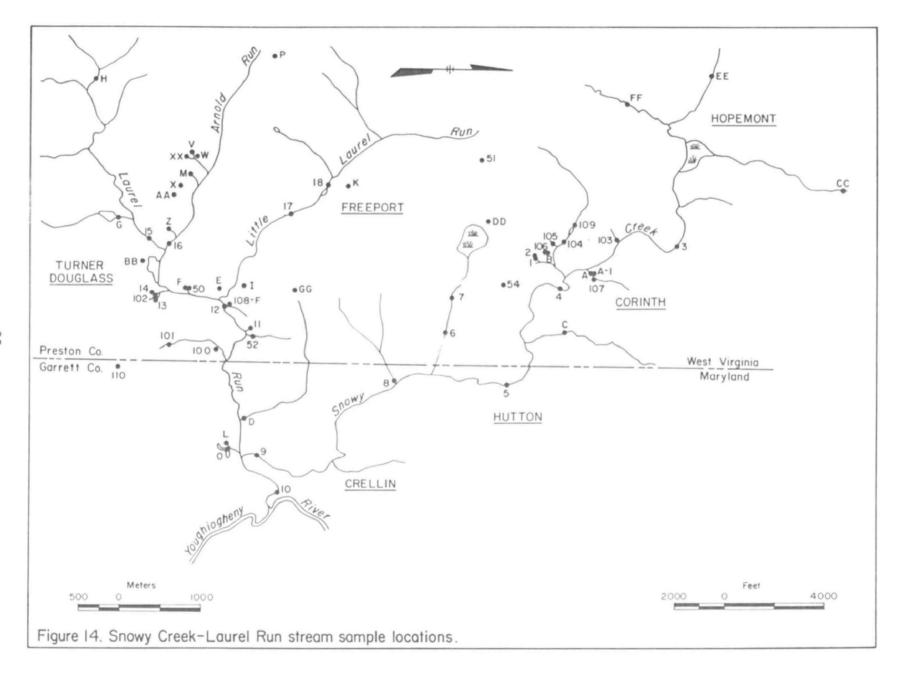
Water samples from stations were grouped into four categories; weekly, bi-weekly, monthly, or supplementary (grab) samples. Samples collected weekly monitored discharges having major impact on Snowy Creek. Stations sampled bi-weekly were either of secondary importance as pollution sources or were stations established to survey the entire drainage basin. Monthly water quality samples were collected in the headwaters of Snowy Creek and Laurel Run. Additional supplementary samples were taken at random throughout the basin. Data obtained during the study are recorded in Appendix C and will be reviewed later in the report.

Flow measurements were recorded at all pollution sources and major stream sampling stations. Measurements were obtained at stream sampling stations using a Gurley pygmy-type current meter to measure the velocity through a known cross section of the stream. Bridges and culverts were used where possible to establish a constant cross section. After periods of high water, cross sections were revised as required. Bench marks were



Source: National Oceanic and Atmospheric Administration, 1951-1974 Precipitation Data

Figure 13. Annual precipitation probability for the Terra Alta, W.Va. weather station.



established at each major stream section, and a gaging system was developed. Data were collected for approximately 5 months at these stations, and stage-discharge curves were plotted. Periodically the stage-discharge curves were checked and revised as necessary. Flows obtained at the mouth of Snowy Creek were correlated with the nearest U.S. Geological Survey gage (number 03075500) on the Youghiogheny River. Details of this correlation are presented later in this section under Hydrology.

Measurement of some discharges was obtained using a V-notch weir. These weirs were installed with the 45° bevel facing upstream as shown in Figure 15. This modified construction has been found to pass more flow at lower heads than a standard V-notch weir. The modified design also provides better nappe separation for monitoring low flows. The comparison of the modified construction curve with the standard construction curve is seen on Figure 16. The modified weirs were calibrated using a bucket and stop watch to measure the weight of water over the weir in a given amount of time. Plotting the relationship of head over the weir and discharge yielded a weir calibration curve.

A continuous monitoring station was also constructed at the mouth of Snowy Creek just above its confluence with the Youghiogheny River. The station is at the Underwood Road Bridge about a fourth of a mile south of Crellin, Maryland. A monitoring unit recording pH and specific conductance on a continuous basis was installed. Figure 17 is a plan of the installation of the monitoring equipment. A stilling well for the station was constructed as shown in Figure 18 to allow the use of a Stevens A-35 stage level recorder. Data collected at this station were used primarily to complement the laboratory analyses and to check flow measurements of unrecorded extremes that might have occurred during the sampling program.

Sampling Program Results

To tabulate and analyze the large quantity of sampling data generated, two computer programs were prepared for use on General Automations Computer Model 18-30. The first Acid Mine Drainage Metric Version (AMDMV) was developed to express output in metric format. The second Acid Mine Drainage English Version (AMDEV) was prepared to express data in English format.

Sample information collected was continuously summarized on a sample run or individual sample location basis. The results of the sampling program are presented in AMDMV format in Appendix C. Appendix C is divided into three types of sample summaries: major sample locations, supplementary (grab) samples and comprehensive water quality analysis summaries. Sample locations are identified as a four digit character such as 0017. The zeros preceeding alpha or numeric characters are dropped in subsequent text references (0017 is reported as 17).

Graphic presentation of flow data collected is outlined in Figure 19. Sample stations of weekly or bi-weekly importance are presented in this figure and in following distribution diagrams. Data presented in Figure 19 indicates that the mean discharge for the Snowy Creek Basin was $128.2 \, \text{m}^3/\text{min}$ (75.5 cfs) as measured at Station 10. Sixty-eight percent or $86.6 \, \text{m}^3/\text{min}$

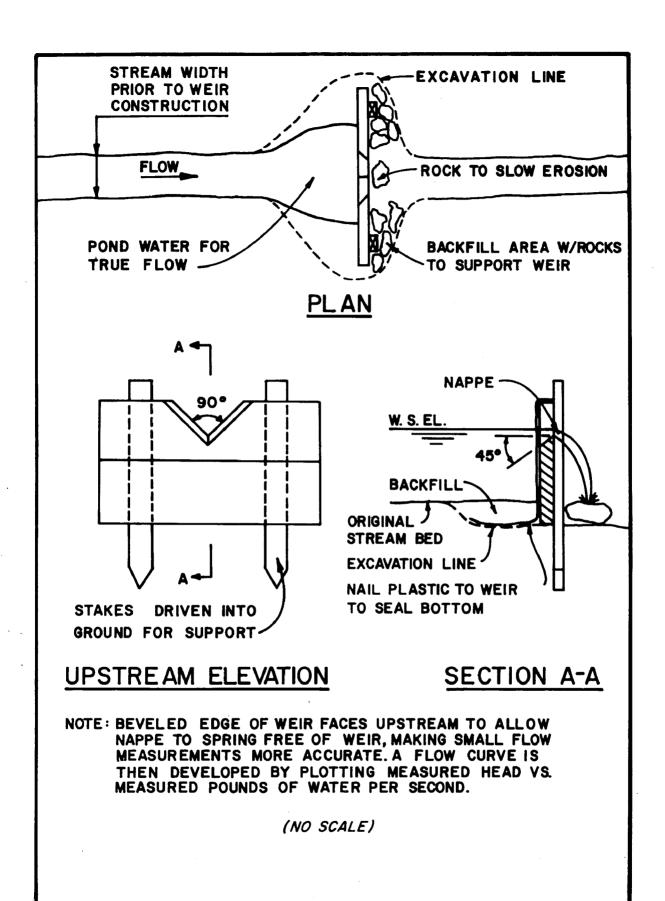
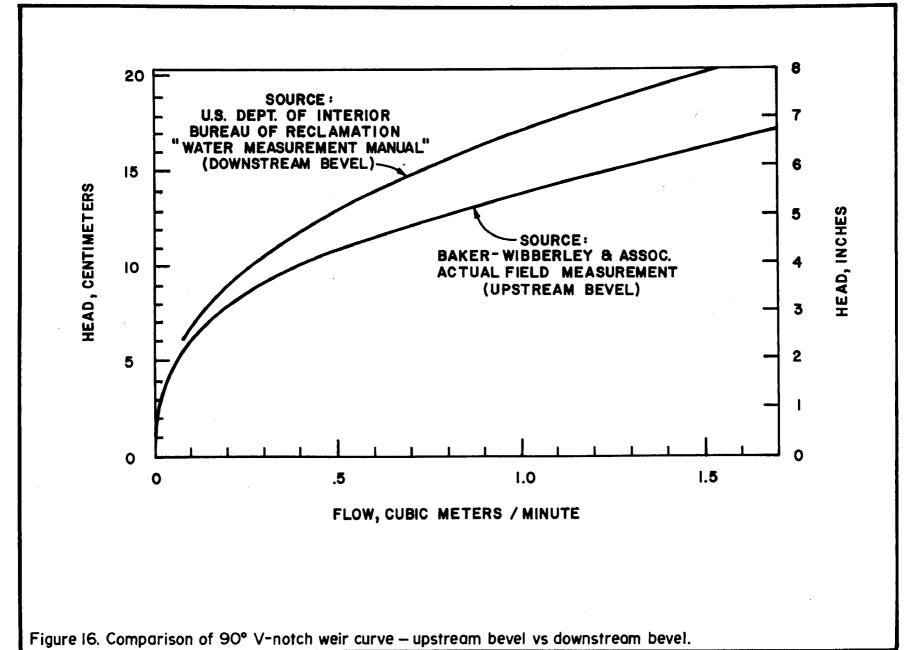
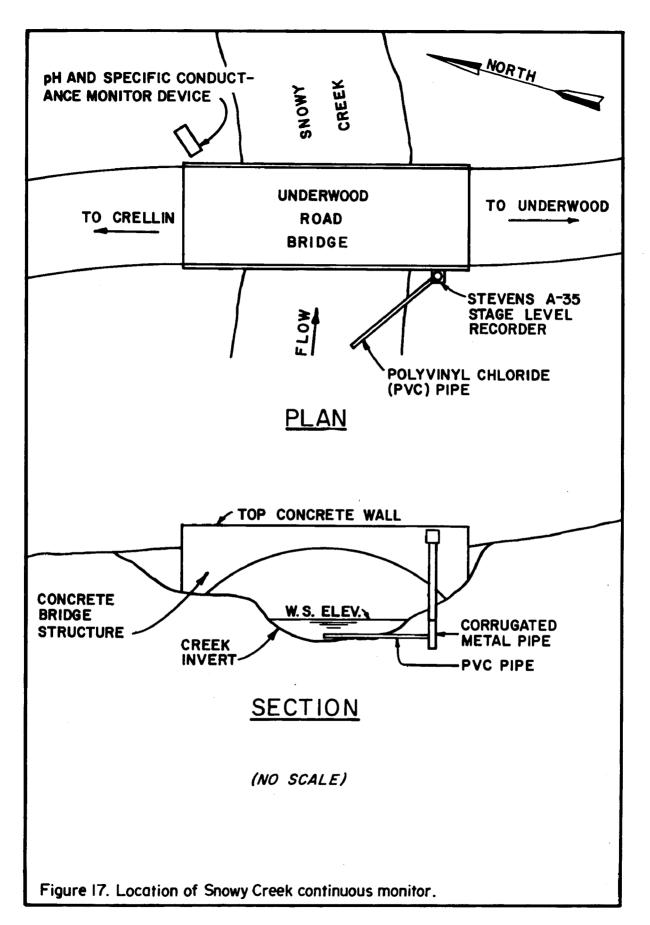
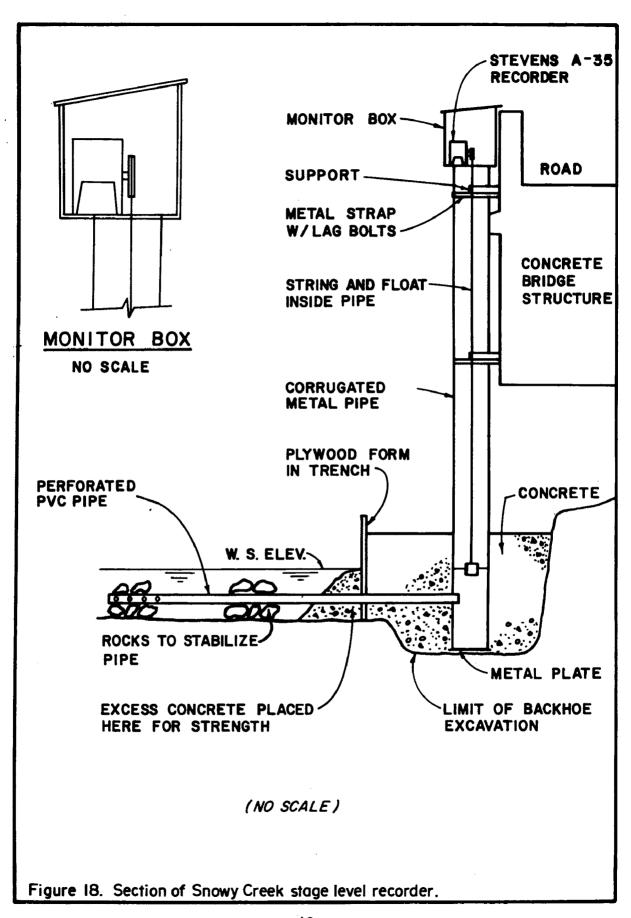
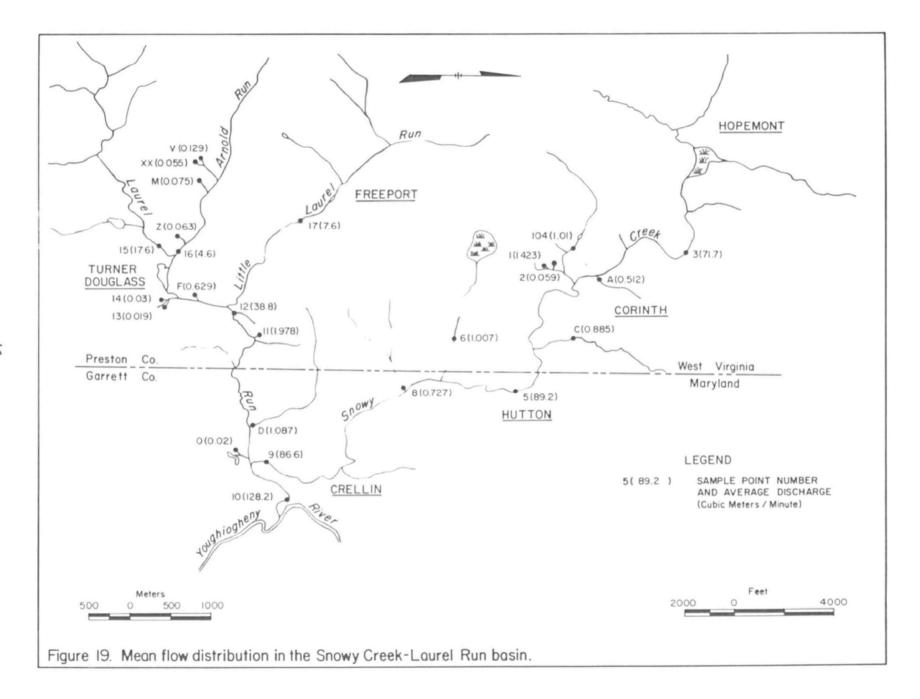


Figure 15. V-notch weir used to measure small flows.









m³/min (51 cfs) of this total is referable to the portion of the basin drained by Snowy Creek and is measured at Station 9. The remaining 41.6 m³/min (24.5 cfs) is contributed by that portion of the basin drained by Laurel Run. Average flow measured at Station 3 was 71.7 m³/min (42.2 cfs) or 56 percent of the total basin flow. Station 3 monitors water quality just above the first acid mine drainage discharge into Snowy Creek. Laurel Run enters Snowy Creek between station 9 and station 10, just before Snowy Creek's confluence with the Youghiogheny River.

Sulfate loadings increase significantly as Snowy Creek enters areas of past or present mining activity. Figure 20 shows two significant increases. The first occurs between Stations 3 and 5, where loadings increase by 128 percent. This result is attributable to the Lima Mine, which is located between these two stations. The second noticeable increase occurs between Stations 9 and 10. An increase in loadings of 132 percent is directly related to the dumping of Laurel Run, which drains the Banner Mine area, into Snowy Creek.

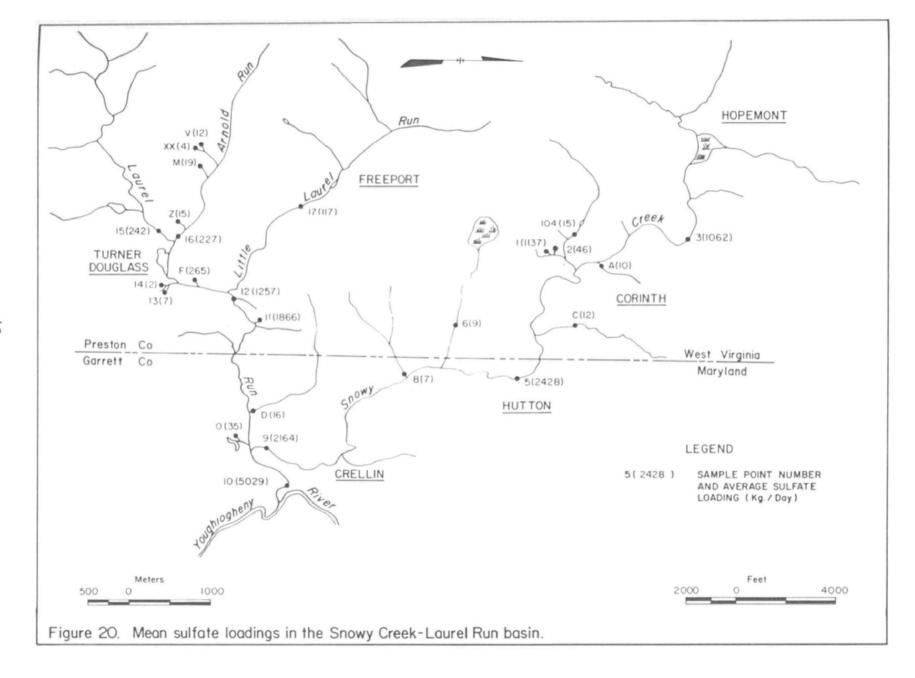
Snowy Creek exhibited a marked net acidity throughout the lower half of the basin (see Figure 21). It has been determined that the only alkaline conditions found in the basin are located upstream of Station 3. Increased acidity occurs between Stations 3 and 5. Introduction of discharges from the Lima Mine area increased loadings from 4 kg/day (10 lb/day), measured at Station 3, to 1,902 kg/day (4,192 lb/day) measured at Station 5. No alkaline conditions were found in Laurel Run. Concentrations of natural acids plus large quantities of acid mine drainage from the Banner Mine complex are the sources of 54 percent of total acid loadings recorded at the mouth of Snowy Creek.

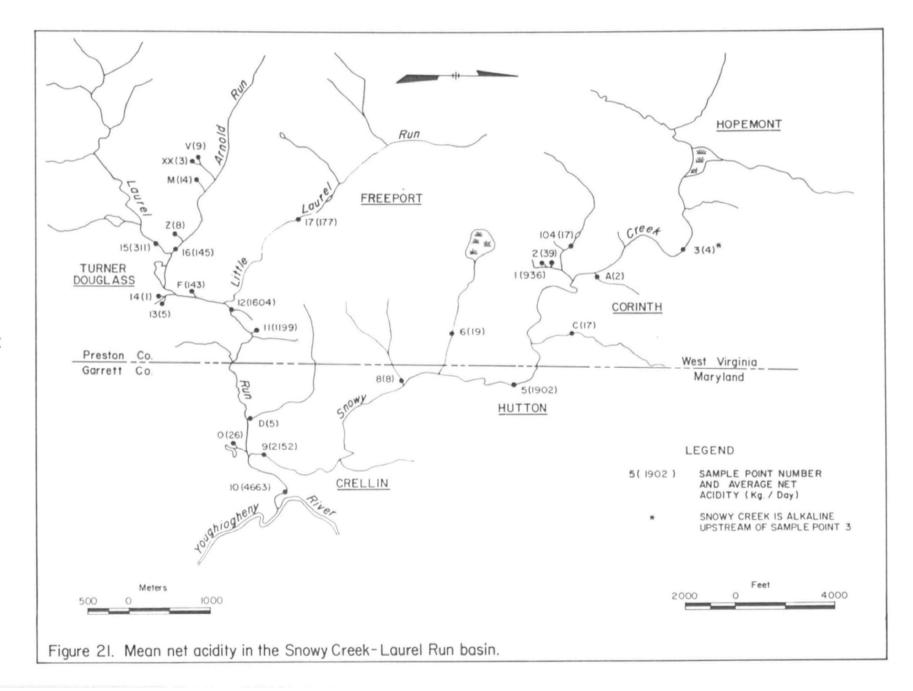
Sampling program results strongly indicate that the Lima and Banner Mine areas are the major sources of acid mine drainage found in the Snowy Creek - Laurel Run Basin.

Samples collected by the Maryland Water Resources Administration (WRA) between 1963 and 1970 indicated that the Youghiogheny River is severely degraded by acid mine drainage from the Snowy Creek Basin. WRA computed stream loadings based upon 1967 to 1970 data (12 sample collections). These calculations show the acid load from the Snowy Creek Basin to vary between 907 and 19,320 kg/day (2,000 and 42,600 lb/day) with an average daily load of 1,364 to 1,814 kg/day (3,000 to 4,000 lb/day) being discharged to the Youghiogheny River. 6

Grab samples taken during the last half of this study sampling program show the Youghiogheny River above Snowy Creek to be alkaline and below its confluence with Snowy Creek to be acidic.

Routine samples collected (38 samples) at the mouth of Snowy Creek (Station 10) during this study, show the acid load contributed by Snowy Creek to the Youghiogheny River ranged between 639 and 25,604 kg/day (1,409 and 56,457 lb/day) with a mean value of 4,663 kg/day (10,282 lb/day).





Hydrology

Flow measurements recorded at Station 10 near the mouth of Snowy Creek were correlated with the U.S. Geological Survey gage located on the Youghiogheny River near Oakland, Maryland. Results of the 38 corresponding measurements are presented in Table 4. On each date that a flow measurement was recorded at Station 10, a corresponding flow measurement was obtained for the U.S. Geological Survey gage. The average flow recorded at Station 10, 128 m³/min (75 cfs), is approximately 26 percent of the average flow, 488 m³/min (287 cfs), recorded at the U.S. Geological Survey gage. The drainage area at Station 10 is 25 percent 88 km² (34 mi²) of the drainage area 347 km² (134 mi²) contributing to the U.S. Geological Survey gage.

TABLE 4.	FLOW	DATA	EVALUATION	
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	Snowy Creek ^a (m ³ /min)	Youghiogheny River ^b (m ³ /min)	
Date sampled	cubic meter/minute	cubic meter/minute	
V 20 1075	77	275	
May 28, 1975	77 83	275	
June 11, 1975	_	230	
June 24, 1975	36	105	
July 7, 1975	61	194	
July 21, 1975	75	105	
August 4, 1975	27	54	
August 18, 1975	296	1,170	
September 2, 1975	468	1,499	
September 16, 1975	66	143	
September 29, 1975	83	362	
October 13, 1975	95	248	
October 27, 1975	77	216	
November 10, 1975	44	104	
November 17, 1975	82	233	
November 24, 1975	66	155	
December 1, 1975	63	148	
December 8, 1975	87	209	
December 15, 1975	139	469	
December 22, 1975	99	308	
December 29, 1975	136	432	
January 5, 1976	309	1,204	
January 12, 1976	92	394	
January 19, 1976	116	332	
January 26, 1976	207	1,244	
Feburary 2, 1976	82	411	
February 9, 1976	102	400	
February 16, 1976	471	1,561	
February 23, 1976	224	896	
March 1, 1976	73	342	

(continued)

TABLE 4 (continued)

Snowy Creek ^a (m³/min) cubic meter/minute	Youghiogheny River ^b (m ³ /min) cubic meter/minute	
29	170	
163	699	
377	1,601	
61	372	
219	1,290	
49	318	
19	145	
- 88	292	
32	<u>216</u>	
4,873	18,546	
128	488	
	(m ³ /min) cubic meter/minute 29 163 377 61 219 49 19 88 32 4,873	(m ³ /min) cubic meter/minute 29 170 163 377 1,601 61 372 219 1,290 49 318 19 145 88 292 32 216 4,873 18,546

a Measured at Station 10. Drainage area 88 m^2 (34 in²).

This comparison has proven valuable for several reasons. First, it serves as a good check for indicating whether open channel flow measurements recorded during the sampling program are of reasonable accuracy. The fact that Snowy Creek does indeed contribute an average of 25 percent of the total flow as measured at the Youghiogheny stream gage can be further used on hydrographs to estimate high flows as recorded by the stage level recorder at the mouth of Snowy Creek.

High flows—those over 1,870 m³/min (1,100 cfs)— as recorded at the Youghiogheny River gaging station for the sampling period are tabulated in Table 5. Dates included are those on which a recorded high flow could have been observed if a routine sample run had been scheduled. Using the 25—percent relationship, a corresponding estimated high flow could be interpolated to have occurred on Snowy Creek. The estimated highest flows probably occurred on January 1, 1976.

Based on historical records at the U.S. Geological Survey gage, Table 6 estimates the magnitude of peak flows and suggested return frequencies of historical floods applicable to Snowy Creek.

Finally, utilization of the 25-percent relationship has developed a quick and accurate method of determining flows on Snowy Creek. Ease of measurement of the estimated flows will be of great value in future monitoring programs on the Snowy Creek - Laurel Run Basin.

b Obtained at U.S. Geological Survey stream gauge. Drainage area 347 km² (134 mi²).

	U.S. Geological Survey* measured flow
Date of occurrence	m ³ /min
August 16, 1975	2,754
August 17, 1975	2,312
August 23, 1975	2,431
August 24, 1975	2,278
December 31, 1975	2,261
January 1, 1976	5,287
January 2, 1976	2,567
January 3, 1976	1,887
February 11, 1976	2,788
February 12, 1976	2,057
February 14, 1976	2,193
February 17, 1976	1,972
February 18, 1976	1,921

^{*} Recorded peak flows not occurring on regular sample runs measured on the Youghiogheny River.

TABLE 6. PEAK FLOWS

	Peak U.S. Geological Survey	Estimated Peak* Snowy Creek
Return frequency	gauge	(m3/min)
2-year	7,361	1,840
5-year	11,271	2,818
10-year	14,161	3,540
25-year	18,190	4,548
50-year	21,250	5,313
100-year	24,140	6,035

^{*} Peak based on a contribution from Snowy Creek of 25 percent of total flow to the U.S. Geological Survey gage.

SOCIAL AND ECONOMIC ENVIRONMENT

Population trends in Preston County have generally reflected the trend of decline in the State's population. A 7-percent decline was recorded from 1960 to 1970 according to published census data. Since 1970 however, there has been an increase of 5-percent in Preston County's population.

The largest town, Kingwood, had a population of 2,550 in 1970. The largest town in the Snowy Creek basin is Terra Alta, with an estimated 1970 population of 1,600.

Major economic activities in Preston County and the Snowy Creek Basin are limited to forestry, agriculture, and mining. At present, probably 50 percent of the county is covered by forest. The forest products industry, however, is still very small. Agricultural activities are primarily limited to small beef and dairy farms. Mining and related industries are major employers in Preston County. 10 As of October 1975, Preston County had more active strip mines than any other county in West Virginia. Because the county is in a mountainous region, tourism may be an important economic contribution for the future. At present there are two State Parks, Cooper's Rock and Cathedral.

Water requirements in the Snowy Creek basin are at present very small. The largest town, Terra Alta, has its own reservoir to supply present and future needs. Downstream on the Youghiogheny River, the town of Oakland, Maryland, uses an average of 980 m³/day (259,000 gal/day) to supplement its public water supply. Surface water use for recreational purposes is also very limited, primarily due to existing substandard quality caused by acid mine drainage.

In the event that the Youghiogheny River becomes a part of the National Wild and Scenic River System, recreation possibilities and tourism will increase in the area. The State of Maryland has already classified the Youghiogheny River as a member of the State Scenic River System. Garrett County, Maryland, according to its development plan, has already established the Maryland sections of the Youghiogheny River, Snowy Creek, and Laurel Run as areas to be utilized for "open space" and "public recreation" needs. Therefore, proposed abatement in the Snowy Creek basin would increase recreational possibilities for both Preston County, West Virginia, and Garrett County, Maryland. 11

SECTION 6

PRELIMINARY ENGINEERING

ABATEMENT METHOD DESCRIPTION

Only 30 percent of the Snowy Creek - Laurel Run Basin is affected by acid mine drainage. Two large pollution sources are responsible for 88 percent of the total acid load discharged from the basin. The two areas, the Lima Mine and the Banner Mine, are the prime targets of major abatement projects as outlined in this study.

Abatement designs in the report have therefore been developed to handle these major pollution sources. Other areas in the study basin exhibit some mine drainage problems, but when compared with the amount of acid mine drainage loadings of the Banner and Lima Mines, their impact on Snowy Creek is negligible.

The most effective abatement measure for the Snowy Creek Basin was determined to be a plan that eliminated the more significant problem areas.

The abatement plans are essentially two separate designs for two individual problem areas. Each design relies on implementation of the other to achieve a total of 75 percent or higher abatement for the Snowy Creek basin. Each design will stand by itself, achieving a percentage of the abatement plan total.

The basic concept of both abatement designs is to inundate additional deep mine workings in the Banner and Lima Mines. With inundation at higher levels, available oxygen will be prevented from reacting with a large amount of acid-producing material, and the circulation and internal mixing that is now present will be eliminated once the mine pools have been stabilized.

Inundation at the Lima Mine should be accomplished by the use of a continuous clay core dam. The Banner Mine will be flooded using an earth dam to create a mine pool level control lake.

Depending on the condition of the barrier remaining between the Lima and Banner Mines, the continuous clay core dam may require extension or reduction in the Freeport area.

PRELIMINARY DESIGN

The Lima Mine Abatement Plan

Presently, only 14 percent of the Lima Mine is flooded. The main discharge point, at an elevation of 743 m (2,438 ft) delineates an approximate shoreline of the present mine pool. The existing mine pool is controlled by the elevation of the existing outfall. Water quality at the main discharge point is not within the acceptable standards for mine drainage discharges as established by the West Virginia Department of Natural Resources.

Data obtained during the sampling period at Station 1, the main discharge point for the Lima Mine, indicate an average flow of $1,423 \text{ m}^3/\text{min}$ (0.837 cfs), an average acid concentration of 457 mg/1, and an average loading of 936 kg/day (2,063 lb/day).

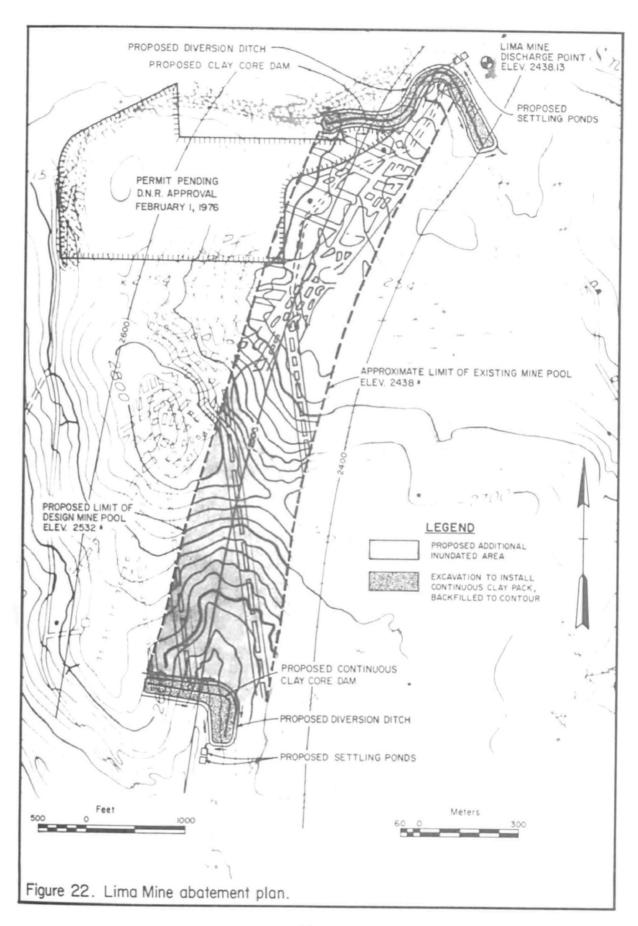
According to stream balances Snowy Creek is alkaline above the Lima Mine area as measured at Sample Station No. 3. There is no acid mine drainage above this point. Once Snowy Creek enteres the Lima Mine area it now becomes an acidic stream picking up approximately 1,859 kg/day (4,099 lb/day). Detailed sample analyses are found in Appendix C.

It has been concluded that the only sources of acid mine drainage entering Snowy Creek upstream from Laurel Run are those in the Lima Mine area. Loadings attributable to this area account for 86 percent of the Snowy Creek total acid load above Laurel Run.

The Design--

The abatement design proposed for the Lima Mine employs a continuous clay core dam to be constructed around the perimeter of old deep workings to an elevation of 772 m (2,532 ft). The clay core will function as an inverted dam with further inundation of the old deep workings. Figure 22 illustrates the role of the continuous clay core dam. The shaded area between the shoreline of the existing mine pool at an approximate elevation of 743 m (2,438 ft) and the shoreline of the proposed mine pool at an approximate elevation of 772 m (2,532 ft) represents the additional deep mine workings to be flooded by the abatement plan. The clay core dam will function both as a mine seal and an artifical barrier to separate mine water.

Excavation to install the continuous clay core dam would be accomplished in much the same manner as excavation to remove the overburden in a stripping operation. A single box cut would be taken along the perimeter of the deep-mine workings, exposing both the coal seam and related workings. Excavation at the southern perimeter (Freeport side) would resemble a large trench. On the northern perimeter (Corinth side), the eastern half of the box cut would also be a large trench, with the western side excavated in the form of an open-sided pit.



Once the earthwork has been completed and the removal of the coal accomplished, the compacted clay core can be constructed as illustrated in Figure 23 and Figure 24. Figure 24 indicates the trenching excavation that will be required to install the clay core on the Freeport side and on the eastern half of the Corinth side. The remaining earthwork configuration will resemble Figure 23.

Installation of a continuous clay core is critical for the proposed abatement plan. In areas where old deep mine workings are encountered, the openings should be packed with impervious material to a minimum of three times the widest opening. The clay core should also be compacted along the entire length of the exposed highwall. The continuous clay core should be at least 3.05 m (10 ft) above the top seam of coal. Figures 23 and 24 indicate the configuration of the proposed clay core.

The clay to be used in the core will be found locally and will be encountered in the excavation for the seal. The clay was sampled and tested by Pittsburgh Testing Laboratory, Pittsburgh, Pennsylvania, and was found to have a permeability of 6.869×10^{-7} cm/sec. The permeability of this local clay is considered suitable for use in the clay core dam.

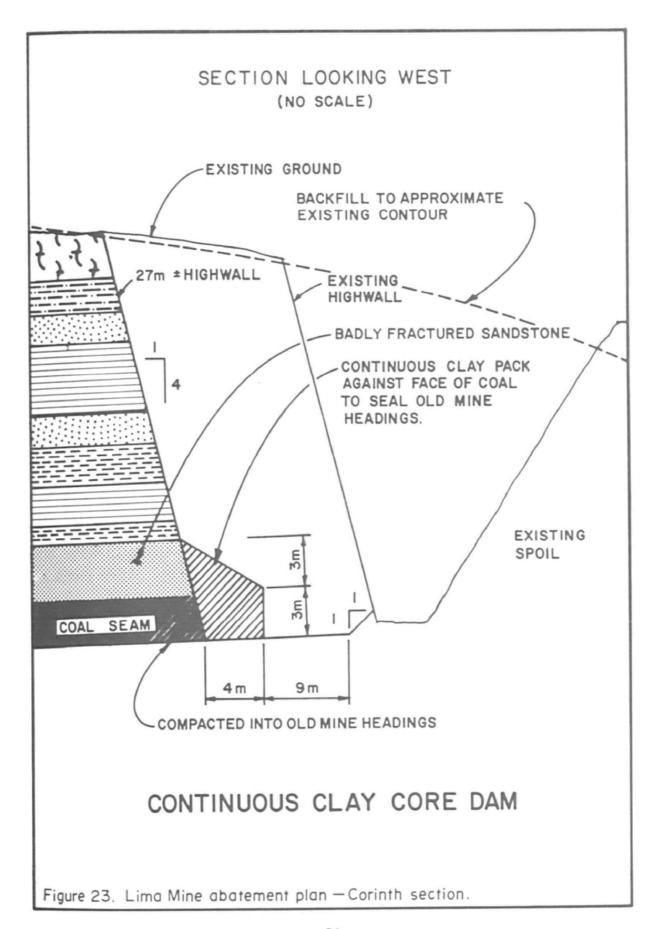
Backfilling will begin once the seal has been installed and will approximate the original contour. The area should be fertilized, mulched and seeded in accordance with recommendations of the U.S. Soil Conservation Service and the West Virginia Department of Natural Resources. Tentatively, a mixture of weeping lovegrass, birdsfoot trefoil, and Kentucky 31 tall fescue is suggested.

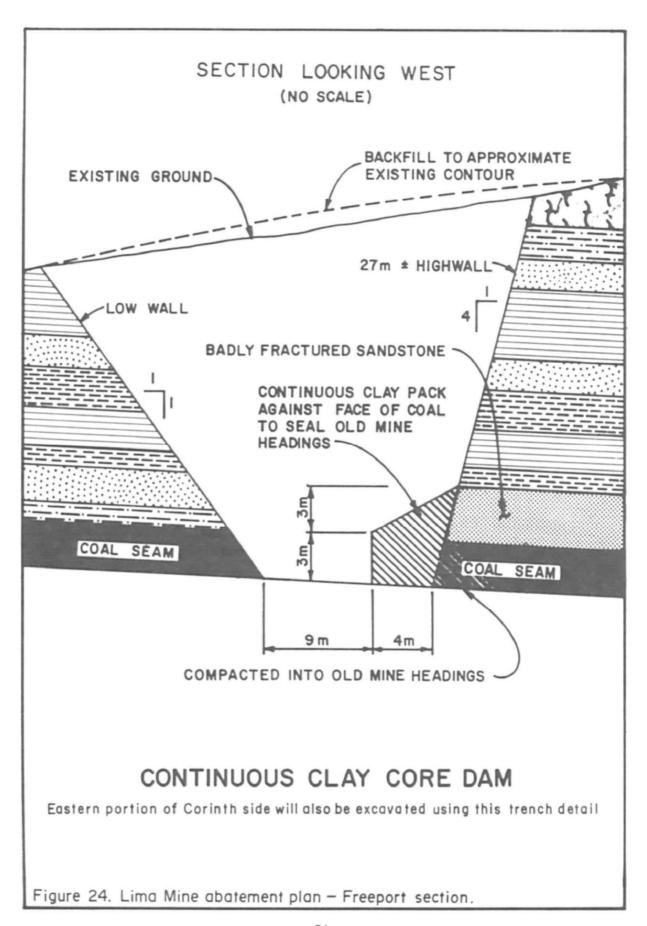
It has been determined that the active stripping operation will reclaim the area adjacent to the proposed clay core dam. Old mine workings and mine dumps will be regraded and revegetated as a part of their present permit.

Conclusions--

Once the clay core dam has been completed it will function in two ways. First, additional flooding of the Lima Mine will inundate approximately 36 percent more of the workings now producing acid mine drainage. Second, the clay core will act as a diversion dam, diverting runoff from the active workings away from the old deep workings. Active surface operations could extract coal above the shoreline of the design mine pool at an elevation of 772 m (2,532 ft). If new stripping permits are issued specifications and requirements to continue the clay core around the perimeter of the design mine pool must be added on all first cuts and on all down-dip ends of the operation joining existing successive segments of the clay core.

If future mining is permitted, eventually the coal and mine workings near the top of the hill above a structure contour of 772 m (2,532 ft) would be daylighted. The recovery of a valuable resource will then be accomplished, and total abatement of the Lima Mine completed.





Failure of the proposed clay core dams is remote. Trench section excavation (Figure 24) will be restrained by the 18 to 24 m (60 to 80 ft) height of the backfill material placed between the undisturbed trench walls. Adjacent to the active surface mine (Figure 23) the clay core dam will be restrained by the 18 to 24 m (60 to 80 ft) of the backfill material placed between the highwall and the existing spoil. Also the minimum width (to original contour) of existing restraining spoil will be approximately 75 m (246 ft) perpendicular to the highwall.

Clay core dams as proposed herein have been used by the mining industry in Pennsylvania for many years. In Somerset County, Pennsylvania the Glessner Mines, Scurfield Coal Company Inc. and Croner Inc. have used this type of clay core dam for over eight years to separate and retain mine water. At the Glessner Mines, the dams have been so successful that tracing agents would not pass through the cores. Scurfield Coal Company Inc. and Croner Inc., have used this type of clay core to seal off abandoned mine workings from their active surface mine operations and have had excellent success.

The Banner Mine Abatement Plan

The Banner Mine and its immediate area have been the principal source of acid mine drainage entering Snowy Creek.

Deep mining has caused many problems and has marred the landscape. At Turner Douglass, the location of the Banner Mine portal nearly 1,007,000 m³ (1,317,156 yd³) of mine waste, with a pH ranging from 3.0 to 3.5, apparently is a major source of acid production. Extensive subsidence resulting from the underground mining has created a swampy area of approximately 42.5 ha (105 acres). The existing swamp, with an elevation of slightly less than 732 m (2,400 ft), is located atop the existing mine pool which has an approximate elevation of 729 m (2,393 ft). Laurel Run, which drains this section of the Snowy Creek basin, meanders through the mine waste and swamp only a few feet above the mine pool.

A series of boreholes drilled when the Banner Mine was in production was used to dewater the mine. Today, with partial inundation of the deep mine, an artesian discharge exists at one of the holes. The borehole is located near the Maryland-West Virginia border and is presently dumping highly acidic waters of the mine pool into Laurel Run after circulation through workings and the mine pool to a depth of about 45 m (148 ft). The deep circulation prohibits stabilization of the pool. This borehole controls the level of the existing mine pool. Earlier attempts to seal this discharge were not successful.

Based on water quality data obtained during the sampling program the Banner Mine can be held responsible for approximately 2,169 kg/day (4,772 lb/day) of acid production. This total was derived by subtracting loadings coming into the Banner Mine complex at Stations 13, 14, 15, 16 and 17 from Station 12. This total is then assumed to be the acid load picked up by

Laurel Run as it passes through the Banner Mine area. Addition of the borehole discharge at Station 11, which is below 12, yields the probable total loading created by the Banner Mine area. Detailed sample analyses for these stations can be found in Appendix C.

There are several additional, but minor, acid mine discharges in the Freeport - Turner Douglass area. These minor discharges are located in the Arnold Run, Freeport, and Kildow mine areas. These discharges are extremely small when compared to the discharges of the Banner Mine complex.

The Arnold Run area at present contributes approximately 145 kg/day (319 lb/day) of acidity, which was measured at Station 16. Actual loadings in the Freeport area were not ascertained, but acid mine drainage discharges are very minor. Loadings created are primarily in response to surface runoff over the mine waste. The problem here is the need for aesthetic improvement. The two openings of the Kildow Mine were monitored at Stations 13 and 14. An average loading of 5 kg/day (11 lb/day) was recorded at Station 13 and an average loading of 1 kg/day (2.2 lb/day) was recorded at Station 14.

The prime goal of the abatement project for the Banner Mine and other outlying areas is a plan that will include elimination of pollution from significant quantities of surface mine waste, sealing of boreholes, negation of the effects of subsidence, closing of mine entries, rechannelization, channel lining, regrading, and revegetation. The proposed abatement design will handle the above problem areas quite adequately.

The Design--

To achieve any significant abatement at the Banner Mine the mine pool had to be raised first so that it could stabilize. Conventional designs employing curtain grouting and borehole sealing could be expected to have only limited success in the area that has a present mine pool only a few feet beneath the surface. As a consequence, a mine pool level control lake, employing an earth dam, was developed.

The plan would be to construct an earth-filled dam that would create an impoundment area situated immediately over the existing mine pool. The new impoundment would now become the surface control for the proposed mine pool levels. Flooding the area to an elevation of 750 m (2,460 ft) would eliminate about 90 percent of the Laurel Run acid mine drainage problem.

This proposed impoundment then eliminates the need for the conventional abatement methods of rechanneling, lining, sealing boreholes, curtain grouting subsidence areas, removing and burying mine waste, regrading, and revegetation. These traditional techniques include a measure of risk. The approach employing the earth dam ensures a higher percentage of abatement for the total project and removes areas of doubtful success.

The earth dam would be placed at a strategic narrow site in the Laurel Run Valley about 0.8 km (1.3 miles) into Maryland. Based on mine maps it is concluded that no underground mining will be encountered at the

proposed dam site. However, as in any project such as this, a detailed subsurface exploration will be required during engineering design. The actual location of the dam and the associated impoundment can be seen in detail in Figure 25. The line indicating a topographic contour of 750 m (2,460 ft) will be the shoreline or extent of the mine pool level control lake. The shaded area inside a structure contour of 750 m (2,460 ft) indicates the subsurface flooding that will be accomplished by the impoundment. Seventy-four percent of the Banner Mine and adjacent workings will be inundated by flooding to an elevation of 750 m (2,460 ft).

A number of criteria were considered in selecting the design elevation of the proposed mine pool. It was determined that the proposed Lima Mine pool limits be established at a higher elevation than the Banner Mine pool. By elevating the present mine pools there will be no adverse effects on local ground water supplies due to the perched water table found in the basin. The design pool at the Lima Mine can be raised to approximately 772 m (2,532 ft) and the Banner Mine pool is designed to an elevation of 750 m (2,460 ft). There is a possibility that the two mines might be interconnected at some elevation above 743 m (2,438 ft).

An elevation of 750 m (2,460 ft) was selected as the maximum lake elevation that could be justified by the size of the contributing water-shed. This elevation will not endanger the existing mill owned by the E.C. Grimm Lumber Company.

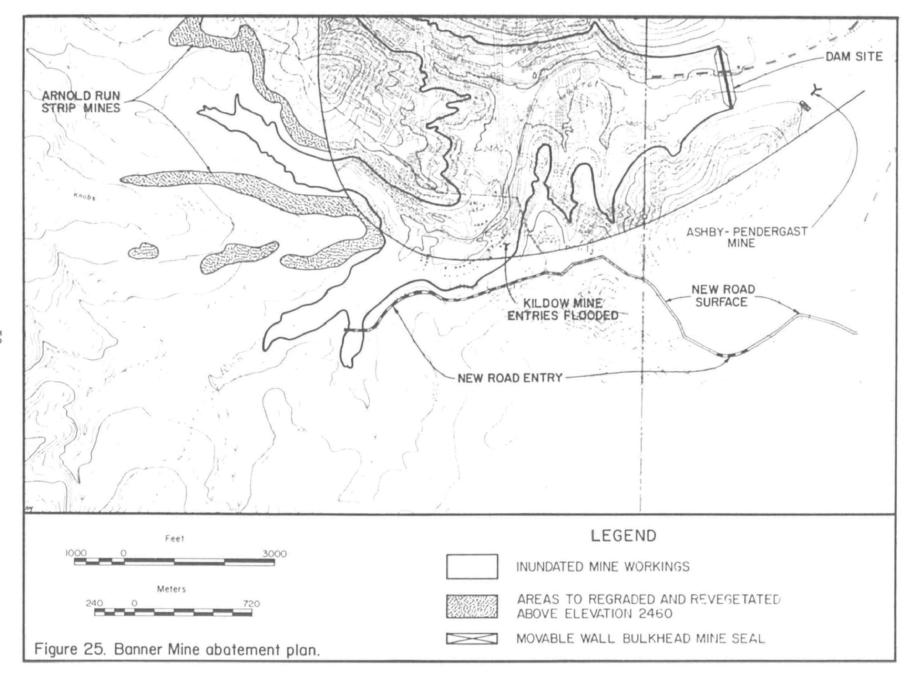
In flooding the Turner Douglass area, only limited amounts of personal property are required. There are only 13 houses located in the impoundment area, most of which are low-cost homes.

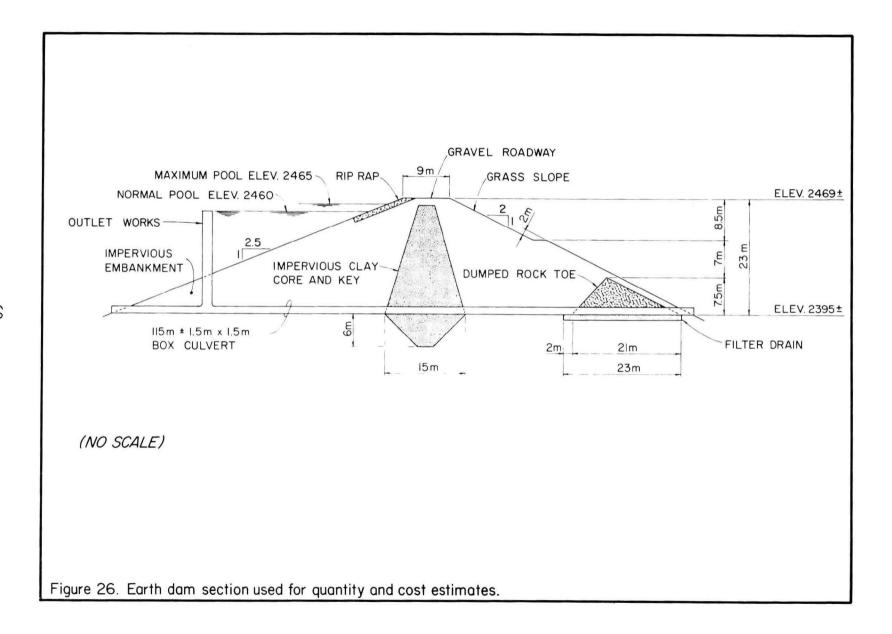
The proposed impoundment limits access for only a few persons. An access road and right-of-way has been included as part of the abatement project to service properties whose present access would be lost by the lake.

The construction of an earth dam (Figure 26) will create the mine pool level control lake. Figure 26 has been developed for preliminary design and cost estimating purposes only. The typical dam section used in cost estimates was derived from the Bureau of Reclamations manual "Design of Small Dams." Detailed design and an onsite evaluation must be conducted during the engineering phases for construction. The estimates obtained using the preliminary design criteria, however, are considered to provide a valid cost analysis for the proposed project.

In the flooding of the Banner Mine area, the possibility of a structural leak exists at the Ashby-Pendergast mine (Figure 25). It is possible that the Banner Mine may have cut into the Pendergast Mine, but this is conjecture. Bulkhead seals will be required at the Ashby-Pendergast Mine to withstand about 29 m (60 ft) of head. A proposed design for a movable wall bulkhead seal is included as Appendix D.

A dewatering borehole exists on Snowy Creek in Maryland at 4,362,229 m north and 630,847 m east (Universal Transverse Mercator Projection). The





borehole is presently not discharging water and is plugged with caved material. However the surface elevation is 733 m (2,405 ft) which is below the elevation of the proposed control lake. This hole must be sealed with a concrete plug before raising the Banner Mine pool.

The remaining areas requiring abatement as considered by the proposed design would include only those locations situated above the level of the control lake, such as the Arnold Run strip mine area and the Freeport mined area. It is important to note here that the impoundment created by the proposed earth dam should inundate the Kildow Mine entries at an elevation of 747 m (2,452 ft).

Abatement in the Arnold Run area would include a standard regrading, construction of diversion ditching, and revegetation done primarily to limit the amount of water introduced to the old spoil area. The Freeport area is also a regrading and revegetation scheme aimed primarily at improving the appearance of the area that lies immediately adjacent to the proposed control lake.

Alternative Design--

As previously mentioned, the conventional approach of rechanneling and lining Laurel Run, sealing boreholes, curtain grouting subsidence areas, removing and burying mine waste, regrading and revegetation might also be used to abate acid mine drainage in the Banner Mine area. But this approach was rejected because of the complex conditions that render the abatement possibilities very questionable.

The alternative design considered an individual abatement method for each problem area and includes subsidence area grouting in the Turner Douglass area, borehole sealing, rechanneling and lining Laurel Run over the Banner Mine, removing and burying of the mine waste at Turner Douglass, and regrading and revegetation in the Turner Douglass, Freeport, and Arnold Run areas.

The alternative design addressed each problem area from the standpoint of increasing the abatement percentage for the Laurel Run area as a whole. An extensive curtain grouting program would be needed to seal the main Banner Mine in the area immediately adjacent to the subsidence-formed swampy area. But such a program would require some knowledge of the number of blind seals needed to guarantee effective inundation of workings. Abatement estimates using this approach are extremely questionable.

The alternative design also proved to be the most costly abatement approach. A complete summary of cost is presented in the following section on capital and operating cost.

Conclusions--

Because the percentage of abatement would be relatively certain, the design approach is the recommended abatement method. The project basin will exhibit a better quality of discharge utilizing the recommended design, and the aesthetic appearance of the project area will be enhanced.

The recommended design is also the least expensive, based on the preliminary cost analysis, presented in the following section.

CAPITAL AND OPERATING COSTS

The following presents estimated abatement costs for the Lima and Banner Mine abatement plans. In presenting the Banner Mine cost estimates, a detailed analysis of the proposed design and the alternative design are included.

Lima Mine Abatement Cost

The cost analysis presented (Table 7) is based on the type of construction and equipment found available in an active surface mining operation. The largest single cost item in the abatement work proposed for the Lima Mine is the excavation of approximately 419,648 $\rm m^3$ (548,900 $\rm yd^3$) of overburden. This large amount of earthwork is required to install the continuous clay core dam.

Items such as diversion ditching, settling ponds, lime application, and swale ditching are included to handle runoff during the construction phases of the abatement work. All construction will conform to existing State and Federal guidelines pertaining to this type of work.

In excavating the overburden to install the continuous clay core dam, a sizable amount of marketable coal is likely to be recovered. Proceeds from its sale could be used to reduce the total project cost.

Recovery figures presented here are based on at least 25 percent of the coal remaining on the Corinth side and 20 percent of the coal remaining on the Freeport side. The percentages used as recovery figures are based on the existence of past underground mining activity. Therefore actual coal recovered may vary according to the extent of underground mining. The recovery of coal could lower the project cost by greater than 40 percent.

Banner Mine Abatement Cost

Cost estimates have been developed for both the design and the alternative abatement plans at the Banner Mine area.

The Design--

Abatement costs covered by the proposed design (Table 8) addresses needs in four main areas. The main Banner Mine complex is handled by the proposed earth dam creating the mine pool level control lake. The closing of the Kildow Mines will also be completed by the impoundment inundating to an elevation of 750 m (2,460 ft). The remaining areas above the impoundment, the Arnold Run strip area and the Freeport area, are covered by a limited program of regrading and revegetation.

TABLE 7 LIMA MINE ABATEMENT COST ESTIMATES

Item	Amount
Drilling, 30,980 m @ \$2.34/m	39,300.00 302,100.00 4,400.00
Settling ponds, 64 dozer hr. @ \$57.75/hr Swale ditching, 1,082 m @ \$3.61/m Lime application, Lump sum Move and install	3,900.00 2,200.00
Impervious clay, 20,325 m ³ @ \$1.80/m Backfill to contour, 4.7 ha @ \$4,620.00/ha Revegetation, 4.7 ha @ \$1,360.00/ha	21,700.00 6,400.00
Construction cost	 49,300.00
Engineering cost @ 7.9% of total construction cost Administrative and legal contingency @ 5% of total	•
Construction cost	
Total project cost (Lima Mine)	642,600.00
Potential credit for coal marketed during construction, 10,770 metric tons @ \$23.15/ton	
Net project cost (Lima Mine)	\$ 393,300.00

TABLE 8 BANNER MINE ABATEMENT COST ESTIMATES: DAM AND IMPOUNDMENT

Item	Amount
Pervious material, 282,137 m ³ @ \$2.75/m ³ \$	775,900.00
Impervious clay core, hauled in	249,100.00
Rip rap, 3,755 m ³ @ \$24.72/m ³	92,800.00
Dumped rock toe, 13,381 m ³ @ \$8.93/m ³	119,500.00
Filter drain, 4,650 m ³ @ $$16.49/m^3$	76,700.00
Class A concrete, 1,008 m ³ @ $$343.75/m^3$	346,500.00
Class B concrete, for items such as a spillway, 994 m ³ @	
\$274.65/m ³	273,000.00
Topsoil, seeding, 10,869 m ² @ $$2.51/m^2$	27,300.00
Storage shed and	
parking area, lump sum	42,000.00
(Continued)	

TABLE 8 (continued)

Item	Amount
Outlet tower, lump sum	17,900.00
mountings, lump sum	21,000.00
Clearing and grubbing, lump sum	65,900.00
Movable wall mine seals $2.0 \pm 21.000 \pm 0.000$	42 000 00
Borehole grouting lump sum	5 300 00
Borehole grouting, lump sum \$ 2	2 154 900 00
New access road:	., 134, 700.00
Excavation, $45,280 \text{ m}^3 \text{ @ } 4.12/\text{m}^3$	188,800.00
Surfacing (gravel road), 4.8 km @ \$34,440.00/km	165,300.00
Drainage. 171 m @ \$137.60/m	23,500.00
Drainage, 171 m @ \$137.60/m	5,300.00
Seeding and mulching, lump sum	10,500.00
Subtotal \$	393,400.00
Land acquisition:	333,400.00
Land for roadway and impoundment, 270 ha @ \$418.00/ha	112,900.00
Improvements, lump sum	68,300.00
Subtotal \$	181,200.00
Arnold Run strip area (east and west bank):	
Regrading, 39.8 ha @ \$4,400.00/ha	175,100.00
Revegetation, 39.8 ha @ \$1,360.00/ha	54,100.00
Diversion ditching, 4602 m @ \$3.61/m	16,600.00
Subtotal \$	242,600.00
Freeport area:	,
Regrading, 5.9 ha @ \$4,400.00/ha	26,000.00
Revegetation, 5.9 ha @ \$1,360.00/ha	8,000.00
Subtotal	34,000.00
Construction cost	,006,100.00
Construction contingency @ 10% of construction cost	300,600.00
Total construction cost	,306,700.00
	181,900.00
Administrative and legal contingency @ 5% of total	
construction cost	165,300.00
Project cost \$ 3	,653,900.00
Project continues & FW of cont	100 700 00
Project contingency @ 5% of project cost Total project cost (Banner Mine)	182,700.00
Total project cost (Lima Mine)	,836,600.00
Total project cost (Lima Mine)	642,600.00
Total abatement cost (proposed design) \$ 4	,479,200.00
Potential credit for coal marketed during construction	249 300 00
Net abatement cost (proposed design) \$ 4	,229,900.00

The Alternative--

Areas requiring abatement were the same as outlined in the design, however, a different method of abatement at the Banner Mine complex was used. Estimated costs for the alternative design are shown in Table 9.

TABLE 9 ESTIMATED COSTS FOR THE ALTERNATIVE BANNER MINE DESIGN

Item	Amount
Channel relocations, 3,292 m @ $$69.30/m $$ Mine waste burial, 1,006,838 m ³ @ $$1.24/m^3 $	•
Regrading, 30.4 ha @ \$4,400.00/ha	1,248,500.00 133,800.00
Revegetation, 30.4 ha @ \$1,360.00/ha	41,300.00
Blind seals (for subsidence area), 200 seals @ \$10,500.00/seal	2,100,000.00
Movable wall mine seals (Kildow Mine + Ashby-Pendergast	, ,
Mines), 4 seals @ \$21,000.00/seal	84,000.00
Subtotal	3,835,700.00
Arnold Run Strip Area (east and west bank):	
Regrading, 39.8 ha @ \$4,400.00/ha\$	175,100.00
Revegetation, 39.8 ha @ \$1,360.00/ha	54,100.00
Diversion ditching, 4,602 m @ \$3.61/m \$ Subtotal \$	16,600.00
	242,600.00
Freeport area: Regrading, 5.9 ha @ \$4,400.00/ha \$	26,000.00
Revergetation 5.9 ha @ \$1.360.00/ha =	8,000.00
Revegetation, 5.9 ha @ \$1,360.00/ha \$ Subtotal \$	34,000.00
,	34,000.00
Construction cost \$	4,112,300.00
Construction contingency @ 10% of construction cost	411,200.00
Construction contingency @ 10% of construction cost $\frac{1}{5}$	4,523,500.00
Engineering cost @ 5.3% of total construction cost Administrative and legal contingency @ 5% of total	239,700.00
construction cost	226,200.00
Project cost	4,989,400.00
Project contingency @ 5% of project cost	249,500.00
Total project cost (Banner Mine)\$	5,238,900.00
Total project cost (Lima Mine)	642,600.00
Total abatement cost (alternative design) \$	5,881,500.00
Potential credit for coal marketed during construction	249,300.00
Net abatement cost (alternative design) \$	5,632,200.00

SECTION 7

IMPLEMENTATION AND OPERATING PLANS

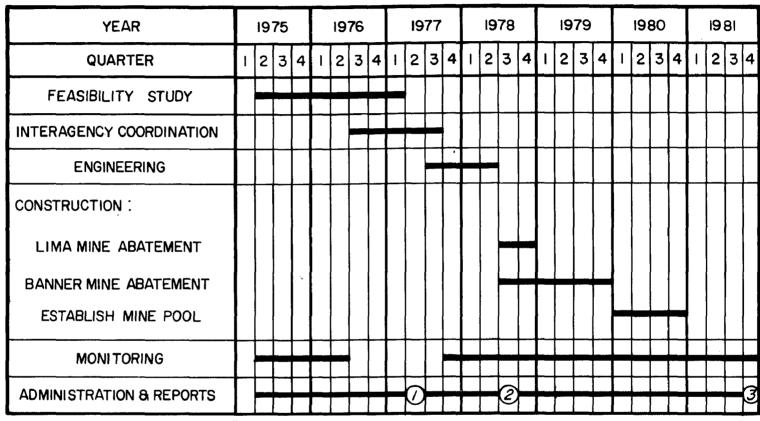
Neither West Virginia nor Maryland alone can implement a feasible successful abatement demonstration project for the Snowy Creek - Laurel Run watershed. Such a project requires either interstate or State-Federal cooperation, or solely Federal involvement.

Since a cooperation effort is needed for the success of the total demonstration project a "lead" agency should be designated as rapidly as possible to evaluate the general involvement of the various government agencies. Screening must be limited to general project scope relative to agency programs. Once the joint venture of government agencies has been assembled, the detailed negotiations between agency programs can be resolved as the other phases of the demonstration project work are being implemented.

Figure 27 presents a timetable of activities that delineates the time periods necessary to accomplish major tasks relative to the proposed abatement demonstration project. This schedule indicates that more than 2 years will be required to perform the design and construction phases of the proposed projects. One year will be required in order to fill and stabilize the proposed lake. In addition, a minimum of 1 hydrological year of monitoring following completion of construction will be required to evaluate the impact of the demonstration project on the water quality of the Youghiogheny River Basin.

Development of engineering plans for construction of the proposed demonstration projects will be in accordance with accepted procedures of the mining industry. Similarly, the U.S. Department of Agriculture, Soil Conservation Service criteria and design considerations will be followed for the development of the engineering plans for the earth structure for the proposed impoundment at the Banner Mine site.

The primary purpose of the proposed demonstration project is to reduce the abandoned mine drainage pollution load being discharged to the Snowy Creek - Laurel Run watershed. Therefore, annual operation of the demonstration project would consist of downstream monitoring of water quality and maintenance of the physical facilities. The proposed Lima Mine abatement is a fixed condition not subject to manipulation. The proposed Banner Mine abatement could be manipulated, but its success is contingent to a major extent on maintaining a nearly constant, static impoundment. Thus operation plans are reduced to routine monitoring and maintenance of physical facilities.



- (1) Final Feasibility Report
- ② Final Engineering Report
- 3 Final Demonstration Project Report

Figure 27. Snowy Creek - Laurel Run project implementation schedule.

SECTION 8

EFFECTIVENESS OF PROJECT

DEMONSTRATION VALUE

The true value of the proposed abatement depends on the effectiveness of the impoundment structures in inundating deep mine workings. In the Lima Mine project, a continuous clay core dam will be used to create an increased mine pool size with a controlled level. This packing may be envisioned as a subsurface trench dam. At the Banner Mine, an earth dam will create an impoundment. The level of the control lake will raise the mine pool and inundate large portions of the Banner Mine workings.

Surface water is now entering the Banner Mine through extensively broken strata in the valleys, especially that of Little Laurel Run. This water now mixes with the waters of the mine pool, circulates through the workings at depth and reports to the surface at the two primary discharge points (Station 11 and F). Establishing the level of the control lake above most points where surface water enters the existing mine pool will eliminate the hydraulic pressure differential responsible for water movement within the mine workings and for the artesian flows at points of discharge. Unpolluted water charging the proposed lake should not become contaminated. Parenthetically, the pH in the underground mine pool will also improve significantly with the stoppage of circulation. In addition, the control lake will inundate mine waste in the vicinity of Turner-Douglass. This action will prevent subaerial acid mine drainage production from the piled and scattered mine waste at the surface.

It is felt that the Banner Mine abatement procedure will provide an effective new approach for eliminating major pollution from favorably situated abandoned mines that have significant fractured overburden and/or surface subsidence.

Implementation of the proposed abatement project at the Lima Mine would reduce acid loadings in Snowy Creek by 1,487 kg/day (3,279 lb/day) for a reduction of about 70 percent. At 80 percent effectiveness of the proposed abatement, the total cost of \$642,600 indicates that each kilogram of acid abated cost \$432 (\$196/lb). This reduction of acid loading would clean up approximately 6.2 km (3.83 miles) of Snowy Creek.

Construction of the proposed design to abate pollution at the Banner Mine should reduce acid loadings from 2,169 kg/day (4,782 lb/day) to 219 kg/day (478 lb/day). Based on an abatement effectiveness of 90 percent,

the total project cost of \$3,836,600 would amount to \$1,967/kg (\$891/1b) of acid removed. Laurel Run would then have 1.6 km (0.98 miles) of cleaner discharge. More important, a new lake would be created with 250 ha (618 acres) of surface area and 17.4 km (10.8 miles) of shoreline. Aesthetic improvement to the entire area both in terms of natural beauty and sanctuary for fish and wildlife will be a definite bonus of the abatement project.

It has been recognized that when impoundments are properly designed and constructed in areas of abandoned mining the resulting ponds can be suitable for supporting and propogating aquatic life. The water quality of the proposed lake should approach values for surface water in the vicinity. It is estimated that the pH will range between 4.6 and 5.4, acidity will be approximately 10-16 mg/l and specific conductivity will range from 50-60 micromhos. These values are similar to values of unpolluted waters as measured at Stations 15 and 16 (Appendix C).

Waters of the above composition will support panfish and other warm water fish. The natural waters of the Laurel Run watershed should not be expected to support fish intolerant to low pH water. Repeated attempts to maintain trout in the natural waters of Laurel Run as found in the pond at Grimm Lumber Company (Sample Station G) have failed. However, bass, sunfish and panfish have flourished.

During construction of the demonstration project, a monitoring program should be implemented during and after abatement to assess effectiveness of the project. Together, both segments of the total Snowy Creek project would cost an average of \$1,200 for each kilogram (\$544/1b) of acid removed.

PUBLIC BENEFITS

The proposed abatement project could eliminate 3,437 kg/day (7,583 lb/day) of acid which is dumped into Snowy Creek - Laurel Run by abandoned mines. This reduction of acid loads to the Snowy Creek - Laurel Run Basin will clean up 6.3 km (3.9 miles) of stream in Maryland and 1.5 km (0.91 miles) of stream in West Virginia. The abatement project will also create a lake, having 17.4 km (10.8 miles) of shoreline and 250 ha (618 acres) of surface area, which will be situated in both Maryland and West Virginia.

With the elimination of 3,437 kg/day (7,583 lb/day) of acid in the study basin, the loads on the Youghiogheny River will also be reduced by an equal amount. The reduced acid loads would now clean up the 7 km (4.4 miles) on the Youghiogheny River, (beginning at its confluence with Snowy Creek and ending at its confluence with the Little Youghiogheny River) recognized by Maryland's Water Resources Administration as severly affected by acid mine drainage from Snowy Creek. Reduced acid loadings would also be carried further downstream on the Youghiogheny River.

Although the lake is primarily a pollution control device, the resulting body of water will definitely have a positive effect in attracting new recreational facilities and providing improved fish and wildlife habitats. The proposed abatement project could fit into the plan of making the Youghiogheny River a National Wild River. The new segment on Snowy Creek and Laurel Run would become another area deserving future preservation under the program.

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APPENDIX A MINE PRODUCTION RECORDS*

TABLE 1-A. MINE PRODUCTION, CORINTH - LIMA MINE AREA

Year	Mine	Owner	Production (tons
1898	Corinth	Oakland Coal +	
		Coke Company	6,330
1899	Corinth	11	8,653
1900	Corinth	11	24,000
1901	1901-1902	11	No record
1903	Corinth	17	29,950
1904	Corinth	11	46,274
1905	Corinth	II .	29,079
1906	Corinth	11	31,919
1907		tt	No record
1908	Corinth	11	20,200
1909	Corinth	11	9,000
1910	Corinth	Jorden Coal Co.	25,780
1911	Corinth	11	18,210
1912	Corinth	11	20,099
1913	1913-1921	10	No production
1921	1713 1721		recorded
1922	Lima #1	Lindsey Coal	12,600
1722	Dime " .	Mining Company	12,000
1923	Lima #1	"	29,954
1924	Lima #1	tt.	3,689
1925	Lima #1	fī	4,052
1926	Lima #1	11	13,419
1927	Lima #1	ri .	53,560
1928	Lima #1 Lima #1	11	62,469
1929	Lima #1	11	39,215
1930	Lima #1 Lima #1	11	
1931	Lima #1	11	11,215
			2,600
1932	1932-1941		No production
1941	Dudu com Date	Posturana Dan	recorded
1942	Princess Pat	Princess Pat	1 (00
70/0	D-1.	Coal Company	1,623
1943	Princess Pat		9,603
1944	Princess Pat		7,765
1945	Princess Pat	"	3,744
1946	Princess Pat	••	5,959
1947	1947–1952	·	No production
1952			recorded
ntinued)			

Table 1-A. (continued)

Year	Mine	Owner	Production (tons)
1953	L+L #1	L+L Coal Co. +	
2,00		Mersing Coal Co.	4,894
1954	L+L #1	n	3,826
1955	L+L #1	11	14,883
1956	L+L #1	IF	61,164
1957	L+L #1	11	67,061
1958	L+L #1	17	17,910
1959	L+L #1	11	1,517
1960	L+L #1	11	885
1961	1961-1974		No mining
1974			3
1975	15-75	Grafton Coal Co.	88,829
Total production			791,930

^{*} Source: West Virginia Department of Mines.

TABLE 2-A. MINE PRODUCTION TURNER DOUGLASS - BANNER MINE AREA

Year	Mine	Owner	Production (tons)
1913	Arnold #1	Kendall Lumber Co.	10,714
1914	ff	11	9,030
1915	**	ff	10,564
1916	**	11	4,000
1917	11	11	5,400
1918	Arnold #1,	**	•
1010	#2 + Guthrie	11	37,310
1919	 11	"	48,397
1920			46,335
1921	17	**	45,316
1922	11	11	23,119
1923	***		33,353
1924	ft .	11	990
1925	Arnold #2	11	10,407
1926	11	ff .	15,154
1927	f1	11	10,501
Total			310,590
1920	Kildow	Kildow Coal Co.	21,728
1921	f 1	tt	12,039
1922	m	fi	8,110
1923	11	ff	13,790
Total			55,667

TABLE 3-A. MINE PRODUCTION FREEPORT - LAUREL RUN AREA

Year	Mine	Owner	Production (tons)
1919	Crane	Freeport Coal Co.	6,000
1920	11	"	51,691
1921	Crane * Kerns	11	40,802
1922	ii Keris	n	30,633
1923	Thayer #1, +		30,033
1,923	#2	n	55,910
1924	ff .		8,805
1925	11	11	17,819
1926	Thayer #1	n	34,512
1927	11	11	23,256
1928	11	Laurel Valley Coal	,
2,00		Company	8,729
1929	ff f	11	41,121
1930	11	11	56,208
1931	11	n	40,111
1932	11	11	
1933	11	**	23,318
	11	**	22,107
1934	"	"	43,187
1935	"		21,580
Total			525,789
1921	Laurel Mine	Laurel Run Coal	
		Company	14,414
1922	11	n	0
1923	11	11	6,702
1924	Fern + Laurel	Fern Coal Co.	5,755
1925	11	11	13,469
1926	n	11	13,264
1927	11	11	979
Total			54,583
1921	Freeport	MDW Coal Co.	10,586
1922	11	**	750
1923	II	11	9,000
Total			20,336
1953	Old Ben	Nordeck Coal Co.	6,577
1954	11	11	5,465
1955	11	11	5,207
1956	11	11	2,846
Total			20,095

TABLE 4-A. MINE PRODUCTION, BANNER MINE

Year	Owner	Comments	Production (tons)
1010	m . n. 1		
1919	Turner Douglass		. 717
	Coal Company		8,714
1920			21,281
1921			0
1922	Oakland Coal Co.		15,622
1923	Stanley Coal Co.		49,176
1924	11	Introduced mining	
		machines	63,919
1925	11		145,649
1926	11		177,210
1927	11	95% Mining machines	207,993
1928	f1	1927-1931 rank	207,707
1929	**	3rd in production	217,164
1930	f1 -	in Preston County	197,741
1931	**	100% mining machines	126,299
1932	***		105,198
1933	11	1932-1940 ranked	116,203
1934	11	lst in production	170,611
1935	11	in Preston County	172,838
1936	11	·	143,708
1937	11		180,578
1938	11	1938 produced 41% of	144,197
1939	11	Preston County total	221,647
1940	11	•	209,637
1941	**	1941 ranked 2nd in	194,100
1942	f1	production in	164,357
_,		Preston County	•
1943	**	,	123,222
1944	**		90,916
1945	11		83,193
1946	**		85,622
1947	f T		91,295
1948	**		94,265
1949	17		141,199
1950	11	Begin final decline	92,010
1951	11	Degin rimi acciano	12,082
1952	**		11,231
1952	11		51,461
	†1		26,413
1954	11		43,347
1955	11	Last year of production	20,970
1956		Last year of production	
Total			4,228,775
		 	

TABLE 5-A. MINE PRODUCTION PRESTON COUNTY

Year	Tons	Year	Tons
1890	159,664	1936	634,467
1894	39,936	1937	596,005
1897	120,211	1938	353,412
1898	169,044	1939	588,701
1899	277,173	1940	784,389
1900	403,610	1941	1,444,752
1901	No Record	1942	1,355,898
1902	No Record	1943	1,808,594
1903	574,741	1944	1,858,175
1904	689,139	1945	1,863,163
1905	651,122	1946	2,476,625
1906	827,772	1947	2,614,357
1907	No Record	1948	2,512,016
1908	874,786	1949	2,388,443
1909	654,333	1950	2,142,305
1910	1,033,902	1951	1,538,846
1911	888,202	1952	1,213,433
1912	841,801	1953	1,228,513
1913	999,141	1954	1,068,265
1914	1,281,181	1955	2,224,145
1915	980,322	1956	1,468,742
1916	1,246,189	1957	2,193,702
1917	1,106,378	1958	1,814,341
1918	1,400,961	1959	2,062,663
1919	1,325,451	1960	2,882,567
1920	1,704,579	1961	2,640,876
1921	1,439,506	1962	2,960,702
1922	939,769	1963	3,089,065
1923	2,182,164	1964	3,240,422
1924	1,668,552	1965	3,922,021
1925	2,733,880	1966	2,188,234
1926	2,165,139	1967	2,717,493
1927	1,970,942	1968	2,239,483
1928	1,921,522	1969	2,415,342
1929	1,934,441	1970	2,470,330
1930	1,600,755	1971	1,906,580
1931	972,289	1972	1,641,960
1932	613,973	1973	1,666,572
1933	597,892	1974	2,822,038
1934	744,699	1975	
1935	736,479		
Total		·	117,509,277

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APPENDIX B METRIC COORDINATES

METRIC COORDINATES

	m.		transverse 7 coordinates (m)	Type of	
Sample No.	Identification	North	East	sample station	Control Control
0001	Lima Mine	4,363,941	629,254	Weekly sample	Double 8" V-notch weir
0002	Lima Mine	4,363,928	629,214	Grab sample	
0003	Snowy Creek (Above Corinth)	4,365,664	629,032	Bi-weekly sample	Stage discharge curve
0004	Snowy Creek (Bridge)	4,364,240	629,585	Grab sample (Monthly)	
0005	Snowy Creek	4,363,609	630,794	Bi-weekly sample	Stage discharge curve
0006	MdW. Va. State Line	4,362,828	630,147	Grab sample (Bi-weekly)	
0007	Swamp Discharge (Bridge)	4,362,902	629,736	Grab sample	36" CMP
8000	MdW. Va. State Line	4,362,216	630,756	Grab sample (Bi-weekly)	
0009	Snowy Creek (Crellin)	4,360,575	631,707	Bi-weekly	Stage discharge curve

Metric coordinates (continued)

Sample No.	<u>mero</u> Identification	universal tr cator zone 17 c North		Type of sample station	Control
0010	Snowy Creek (Crellin) Below Laurel Run	4,360,812	632,156	Weekly	Stage discharge curve
0011	Slime Hole	4,360,410	630,206	Weekly	Double 8" V-notch weir
0012	Laurel Run	4,360,144	629,887	Bi-weekly	Stage discharge curve
0013	Kildow Mine	4,359,240	629,808	Bi-weekly	8" V-notch weir
0014	Kildow Mine	4,359,228	629,759	Bi-weekly	8" V-notch weir
0015	Laurel Run	4,359,132	629,056	Bi-weekly	Stage discharge curve
0016	Arnold Run	4,359,412	629,106	Bi-weekly	Stage discharge curve
0017	Little Laurel Run South of Freeport	4,360,914	628,708	Bi-weekly	Stage discharge curve
0018	Little Laurel Run (Road)	4,361,362	628,358	Grab sample	48" Concrete pipe
0050	Out of GOB (Near OOOF)	4,359,655	629,732	Grab sample	
0051	Grab	4,363,240	628,024	Grab sample	
0052	Grab	4,360,468	630,250	Grab sample	

Sample No.			transverse 7 coordinates (m) East	Type of sample station	Control
0054	Grab	4,363,549	629,556	Grab sample	
0100	Reckhart Hole	4,360,012	630,408	Grab sample	
0101	Reckhart Stream	4,359,448	630,360	Grab sample	
0102	Stream at 0013 and 0014	4,359,225	629,770	Grab sample	8" V-notch weir
0103	•	4,364,928	628,981	Grab sample	
0104	Pond Drainage	4,364,279	629,011	Grab sample	
0105		4,364,137	629,037	Crab sample (Bi-weekly)	
0106	Grafton Strip Drain	4,364,072	629,136	Grab sample (Bi-weekly)	12" Pipe
0107	Mine Pit Drain into 000A	4,364,663	629,465	Grab sample	
0108-F	Swamp Drain into 0012	4,360,171	629,861	Grab sample	
0109	Pond above Grafton	4,364,387	628,817	Grab sample	
0110	Simms Mine	4,358,760	630,660	Grab sample	

	me		transverse 7 coordinates (m)	Type of	
Sample No.	Identification	North	East	sample station	Control
000A	Stream (Corinth)	4,364,619	629,397	Grab sample (Bi-weekly)	
00A1	Swamp Drain into 000A	4,364,648	629,390	Grab sample	
000В	Grafton Settling Pond	4,364,091	629,148	Grab sample	
000C	Stream At Corinth	4,364,300	630,125	Grab sample (Bi-weekly)	36" Concrete pipe
000D	Stream into Laurel Run	4,360,372	631,260	Grab sample (Bi-weekly)	
000E	Yellow Boy BH	4,360,048	629,667	Grab sample	8" V-notch weir
000F	Glory Hole	4,359,643	629,728	Weekly	Double 8" V-notch wei
000G	Grimms Pond Out- fall	4,358,724	628,792	Grab sample	
000н	Headwater Laurel Run at Bridge	4,358,460	627,108	Monthly	Stage discharge curve
0001	Spring	4,360,341	629,626	Grab sample	
000J	GOB Fire	4,359,550	629,503	Grab sample	FO

Metric coordinates (continued)

Universal transverse mercator zone 17 coordinates (m) Type of										
Sample No.		North	East	sample station	Control					
000К	Valley Disch. (Freeport)	4,361,610	628,365	Grab sample (Monthly)						
000L	Pond at 0000	4,360,175	631,602	Grab sample	8" V-notch weir					
000M	Arnold Mine	4,359,662	628,240	Bi-weekly	8" V-notch weir					
0000	Pendergast	4,360,156	631,592	Weekly	8" V-notch weir					
000P	Arnold Run Headwater	4,360,673	626,767	Grab sample (Monthly)						
0000	Arnold Strip	4,359,650	627,988	Bi-weekly	8" V-notch weir					
000W	Strip Outbreak	4,359,739	628,024	Grab sample						
000x	Strip Outbreak	4,359,549	628,388	Grab sample						
000Z	Strip Mine	4,359,400	628,912	Bi-weekly	8" V-notch weir					
00AA	Highwall at Grimms	4,359,472	628,504	Grab sample						
ООВВ	Stream at Turner Douglass	4,359,234	629,650	Grab sample						
00CC	Road to Alpine Lake	4,367,691	628,300	Grab sample (Monthly)	2 - 48" x 76" Concrete pipes					
OODD	Pritt Farm	4,363,336	628,758	Grab sample	. 					
(continued	(continued)									

Metric coordinates (continued)

Universal transverse									
Comple No			coordinates (m) East	Type of sample station	Comerce 1				
Sample No.	Idencification	NOTER	Dabe	Sample Station	Control				
OOEE	Stream to Lima	4,366,051	626,920	Grab sample					
OOFF		4,365,014	627,290	Grab sample					
00GG	Core Drill	4,360,964	629,676	Grab sample					
XX0C	Arnold Strip	4,359,621	628,000	Bi-weekly	8" V-notch weir				
YR01	Yough Below Snowy Creek	4,361,094	632,345	Grab sample					
YRO2	Yough Below Snowy Creek	4,358,096	632,465	Grab sample					
	Bridge at Monitor	4,360,677	632,312	No sample					

APPENDIX C

COMPUTERIZED PRINTOUT* - COMPREHENSIVE WATER ANALYSES

Presented is a detailed summary of all samples collected during the course of the study. The tables presented are from standard IBM computer sheets that have been reduced photographically in order to be included in the report. Comprehensive water analyses are also included in this section.

The chemical analyses of the Snowy Creek and Laurel Run samples for aluminum, calcium, chromium, copper, iron, lead, magnesium, manganese, potassium, sodium and zinc were completed by atomic absorption spectrophotometry. The analyses for ferrous iron, sulfate, hexavalent chromium and fluoride were performed colorimetrically using the reference, "Standard Methods for the Examination of Water and Wastewater," 13th Edition 1971 APHA, AWWA, WPCF. Acidity, total solids, suspended solids, and chemical oxygen demand were determined in accordance with the methods expressed in "Methods for Chemical Analysis of Water and Wastes," 1974, EPA. pH was measured electrometrically. Specific conductance was determined at 25° C by using a Wheatstone bridge. Chloride was analyzed by using the mercuric nitrate titration method found in "Annual Book of Standards, Part 23, Water, Atmospheric Analysis," 1972, ASTM. Turbidity was determined in a turbidimeter and all hardness values reported were calculated in accordance with Standard Methods using the atomic absorption values obtained for Ca, Mg, Fe, Al, Zn, and Mn.

^{*} Headings on computer sheets labeled "IRON (FERR)" refer to ferrous iron.

TABLE 1-C. SAMPLE STATION 0001, LIMA MINE

Analysis a	May 15, 1975	December 8, 1975	May 3, 1976
рH	2.60	2.50	2.90
Acidity	450.00	650.00	464.00
Alkalinity	0.00	0.00	0.00
Iron (total)	62.00	72.00	66.00
Iron (ferrous)	0.50	8.00	11.00
Specific conductance			
(uMhos)	1,350.00	1,350.00	960.00
Sulfate (as SO ₄)	550.00	675.00	580.00
Aluminum	44.00	50.00	43.00
Cadmium		0.002	0.007
Calcium	22.00	26.00	28.00
Chromium (hexavalant)	0.01	0.01	0.01
Chromium (total)	0.01	0.01	0.01
Copper	0.09	0.04	0.10
Cyanide		0.01	0.06
Lead	0.05	0.05	0.05
Magnesium	9.30	11.50	11.50
Manganese	0.68	0.80	1.30
Mercury		ND	0.001
Potassium	1.10	1.30	1.20
Sodium	0.42	0.50	0.47
Zinc	0.69	0.82	0.81
Arsenic		0.011	0.002
Turbidity (APHA units)	0.44	1.30	1.50
Total solids	1,000.00	980.00	916.00
Suspended solids	0.20	0.80	14.50
Total organic carbon	-	0.65	5.90
Chloride	4.00	5.00	5.00
Fluoride	0.65	0.70	0.72
C.O.D.	8.60	6.50	17.00
Hardness	450.00	522.00	479.00

TABLE 2-C. SAMPLE STATION 0011, SLIME HOLE

Analysis	May 15, 1975	December 8, 1975	May 3, 1976
Ph	3.00	2.90	2.90
Acidity	335.00	430.00	368.00
Alkalinity	0.00	0.00	0.00
Iron (total)	93.00	120.00	100.00
Iron (ferrous)	90.00	120.00	96.00
Specific conductance	30.00	120:00	70.00
(uMhos)	1,060.00	1,180.00	1,080.00
Sulfate (as SO4)	595.00	635.00	570.00
Aluminum	29.00	27.00	28.00
Cadmium	29.00	0.002	0.008
Calcium	55.00	53.00	50.00
Chromium (hexavalant)	0.01	0.01	0.01
Chromium (total)	0.01	0.01	0.01
•	0.06	0.03	0.08
Copper Cyanide	0.00	0.01	0.07
Lead	0.05	0.05	0.05
Magnesium	20.00	21.00	24.00
Manganese	2.00	2.10	1.70
Mercury	2.00	ND	0.001
Potassium	3.50	2.90	2.80
Sodium	0.67	0.83	0.71
Zinc	0.79	0.77	0.70
Arsenic	0.73	0.011	0.009
Turbidity (APHA units)	1.30	3.10	2.20
Total solids	1,010.00	976.00	1,000.00
Suspended solids	0.60	0.30	0.20
Total organic carbon		0.45	4.40
Chloride	1.00	1.00	2.00
Fluoride	1.40	1.10	1.20
C.O.D.	18.00	18.00	20.00
Hardness	550.00	589.00	564.00

TABLE 3-C. SAMPLE STATION 0050, BELOW GLORY HOLE

Analysis	December 8, 1975
рН	3.30
Acidity	50.00
Alkalinity	0.00
Iron (total)	0.63
Iron (ferrous)	0.05
Specific conductance	
(uMhos)	280.00
Sulfate (as SO4)	61.00
Aluminum	2.50
Cadmium	0.002
Calcium	3.10
Chromium (hexavalant)	0.01
Chromium (total)	0.01
Copper	0.01
Cyanide	0.01
Lead	0.05
fagnesium	2.60
langanese	0.42
fercury entry	ND
Potassium	0.89
Sodium	0.54
Zinc	0.10
Arsenic	0.002
Curbidity (APHA units)	1.00
Cotal solids	94.00
Suspended solids	0.20
Cotal organic carbon	0.40
Chloride	2.00
luoride	0.20
C.O.D.	2.10
lardness	35.00

TABLE 4-C. SAMPLE STATION 000F, GLORY HOLE

Analysis	May 15, 1975	May 3, 1976
[3.10	3.40
idity	110.00	89.00
kalinity	0.00	0.00
on (total)	12.00	1.90
on (ferrous)	4.70	0.86
cific conductance		
uMhos)	640.00	440.00
fate (as SO4)	255.00	156.00
minum	13.00	10.00
mium		0.007
cium	28.00	17.00
comium (hexavalant)	0.01	0.01
omium (total)	0.01	0.01
er	0.01	0.05
nide		0.03
đ	0.05	0.05
nesium	12.00	9.70
ganese	1.90	1.60
cury		0.001
assium	1.80	1.40
Lum	0.50	0.53
c	0.31	0.25
enic		0.003
bidity (APHA units)	2.60	0.55
al solids	760.00	345.00
pended solids	0.20	6.10
al organic carbon		3.90
oride	7.00	9.00
oride	2.00	1.10
.D.	4.10	5.00
rdness	206.00	145.00

TABLE 5-C. SAMPLE STATION 000J, STREAM FROM GOB FIRE

Analysis	May 3, 1976
рН	2.00
Acidity	4,800.00
Alkalinity	0.00
Iron (total)	2,200.00
Iron (ferrous)	2,200.00
Specific conductance	
(uMhos)	7,000.00
Sulfate (as SO4)	5,800.00
Aluminum	250.00
Cadmium	0.135
Calcium	29.00
Chromium (hexavalant)	0.01
Chromium (total)	0.12
Copper	0.11
Cyanide	0.01
Lead	0.05
Magnesium	120.00
Manganese	2.90
Mercury	0.003
Potassium	82.00
Sodium	35.00
Zinc	2.00
Arsenic	0.379
Turbidity (APHA units)	8.10
Total solids	10,000.00
Suspended solids	5.80
Total organic carbon	48.80
Chloride	39.00
Fluoride	6.50
C.O.D.	453.00
Hardness	5,930.00

TABLE 6-C. SAMPLE STATION 0000, PENDERGAST

Analysis	May 15, 1975	December 8, 1975	May 3, 1976
рН	2.70	2.50	2.60
Acidity	866.00	780.00	952.00
Alkalinity	0.00	0.00	0.00
Iron (total)	158.00	160.00	200.00
Iron (ferrous)	42.00	55.00	65.00
Specific conductance			
(uMhos)	2,060.00	1,780.00	1,860.00
Sulfate (as SO4)	1,225.00	1,170.00	1,290.00
Aluminum	54.00	49.00	60.00
Cadmium		0.002	0.009
Calcium	196.00	59.00	89.00
Chromium (hexavalant)	0.01	0.01	0.01
Chromium (total)	0.06	0.03	0.05
Copper	0.30	0.11	0.25
Cyanide		0.01	0.01
Lead	0.02	0.05	0.05
Magnesium	34.00	36.00	43.00
Manganese	1.60	1.60	1.60
Mercury		ND	0.001
Potassium	1.70	1.20	1.60
Sodium	0.77	0.73	0.61
Zinc	4.20	3.40	3.30
Arsenic		0.005	0.003
Turbidity (APHA units)	1.20	5.10	4.00
Total solids	1,150.00	1,670.00	1,860.00
Suspended solids	6.00	0.50	0.60
Total organic carbon	· 	0.40	2.70
Chloride	15.00	5.00	8.00
Fluoride	1.50	1.10	1.40
C.O.D.	4.20	18.00	13.00
Hardness	1,220.00	858.00	1,100.00

TABLE 7-C. SAMPLE STATION OOOV, ARNOLD STRIP

Analysis	December 8, 1975
рН	2.90
Acidity	80.00
Alkalinity	0.00
Iron (total)	0.48
Iron (ferrous)	.10
Specific conductance	
(uMhos)	390.00
Sulfate (as SO4)	92.00
Aluminum	3.30
Cadmium	0.002
Calcium	6.90
Chromium (hexavalant)	0.01
Chromium (total)	0.01
Copper	0.01
Cyanide	0.01
Lead	.05
Magnesium	3.70
Manganese	0.38
Mercury	ND
Potassium	1.40
Sodium	0.33
Zinc	0.12
Arsenic	0.001
Turbidity (APHA units)	1.00
Total solids	128.00
Suspended solids	0.20
Total organic carbon	0.50
Chloride	2.00
Fluoride	0.25
C.O.D.	4.50
Hardness	53.00

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WEST VIRGINIA ACID MINE DRAINAGE STUDY SNOWY CREEK-LAUREL RUN WATER SAMPLING DATA

LOCATION 0001 LIMA MINE

			-															
		FLOW	PH	SPEC	ACIO	\	ALKALIN	1 T T Y	IRON(1	(JATOT	IRON (FERR)	SUI	LFATE	ALUN	INUM	MANG	ANESE
		CMM		AB COND		KG/D			MG/L		MG/L		MG/L		MG/L	-	MG/L	KG/D
DATE	IDENT	Griff	FIELD (. NO COND	MG/ L	KG/ P	710 / L	K 0 / U	1107 C	NO, 0		,,,,,,						
051475	LIMA MINE	0.523	2.5	2.6 1350.	450.	339.	0.	0.	62.0	46.7	0.5	0.4	550.		44.00	33.2	0.68	0.5
052175	LIMA MINE	0.867	2.5	2.7 1290.	424.	529.	0.	0.	59.0	73.6	1.3	1.6	530.		42.00	52,4	0.69	0.9
052875	LIMA MINE	0.855	2.4	2.8 1040.	334.	411.	0.	0.	39.0	48.0	0.8	0.9	445.		30.00	36.9	0.55	0.7
060475	LIMA MINE	0.867	2.8	2.5 1140.	387.	483.	0.	0.	47.0	58.6	1.4	1.7	530.		35.00	43.7	0.60	0.7
061175	LIMA MINE	0.625	2.5	2.6 1050.	456.	411.	0.	0.	59.0	53.1	1.1	1.0	575.		41.00	36.9	0.75	0.7
061675	LIMA MINE	0.748	2.7	2.6 1160.	425.	458.	0.	0.	58.0	62.4	3.1	3.3	530.		41.00	44.1	0.71	0.8
062475	LIMA MINE	0.471	2.6	2.5 1290.	514.	348.	0.	0.	64.0	43.4	1.2	0.8	635.		49.00	33.2	0.91	0.6
063075	LIMA MINE	0.460	2.6	2.6 1320.	525.	348.	0.	0.	70.0	46.4	1.9	1.3	650.		52.00	34.5	0.81	0.5
070775	LIMA MINE	0.285	2.6	2.5 1500.	670.	275.	0.	0.	72.0	29.6	1.7	0.7	700.		55.00	22.6	1.00	0.4
071475	LIMA MINE	0.267	2.5 - 7	.5 1280.	622.	239.	0.	0.	82.0	31.5	9.5	3.6	720.		56.00	21.5	0.90	0.3
072175	LIMA MINE	0.253	2.7	2.5 940.	440.	160.	0.	0.	68.0	24.8	13.0	4.7	615.		45.00	16.4	0.72	0.3
072875	LIMA MINE	0.260	2.6	2.4 1020.	606.	227.	0.	0.	80.0	29.9	4.6	1.7	718.		53.00	19.8	1.00	0.4
080475	LIMA MINE	0.224	2.7	2.6 1420.	674.	218.	0.	0.	100.0	32.3	31.0	10.0	762.		60.00	19.4	1.20	0.4
081175	LIMA MINE	0.214	2.6	.5 1350.	716.	221.	0.	0.	105.0	32.4	29.0	8.9		270.		19.4	1.40	0.4
081875	LIMA MINE	2.066	2.6	2.6 1230.	597.	1776.	0.	0.		264.8	33.0	98.2		1993.			0.85	2.5
082575	LIMA MINE	2.839	2.7	2.6 1020.	440.	1799.	0.	0.		171.7	0.5	2.0		2024.			0.60	2.5
090275	LIMA MINE	4.910	2.7	2.8 845.	282.	1994.	0.	0.		190.9	0.2	1.1		2213.			0.51	3.6
090875	LIMA MINE	1.009	2.7	2.7 1225.	392.	570.	0.	0.	52.0	75.6	1.2	1.7		724.		52.3	0.68	1.0
091675	LIMA MINE	0.510	2.4	.7 1210.	645.	473.	0.	0.	72.0	52.8	2.5	1.8	625.		45.00	33.0	0.80	0.6
092275	LIMA MINE	0.826	2.7	2.7 1180.	460.	547.	0.	0.	60.0	71.3	10.0	11.9	600.		40.00	47.6	0.69	0.8
092975	LIMA MINE	1.009	2.7 2	.6 1280.	655.	952.	0.	0.	64.0	93.0	1.0	1.5	625.		45.00	65.4	0.79	1.1
100675	LIMA MINE	0.564	2.7	.5 1480.	684.	556.	0.	0.	71.0	57.7	2.0	1.6	735.		52.00	42.2	0.82	0.7
101375	LIMA MINE	.0.693	2.7	2.6 1370.	479.	478.	0.	0.	57.0	56.9	4.6	4.6	610.		44.00	43.9	0.74	0.7
102075	LIMA MINE	1.614	2.5	.7 1350.	589.	1369.	0.	0.		153.4	6.1	14.2		1511.			0.80	1.9
102775	LIMA MINE	0.765		2.5 1380.	680.	749.	0.	0.	70.0	77.1	1.6	1.8	748.		51.00	56.1	0.89	1.0
110375	LIMA MINE	0.629	2.8	2.5 1500.	696.	630.	0.	0.	75.0	67.9	4.5	4.1	775.		55.00	49.8	0.96	0.9
111075	LIMA MINE	0.510		2.6 1280.	659.	484.	0.	0.	70.0	51.4	4.4	3.2	708.		50.00	36.7	0.87	0.6
111775	LIMA MINE	0.765		2.7 1210.	500.	550.	0.	0.	64.0	70.5	8.0	8.8	635.		46.00	50.6	0.79	0.9
112475	LIMA MINE	0.693		2.5 1280.	677.	676.	0.	0.	76.0	75.9	3.0	3.0	742.		53.00	52.9	0.90	0.9
120175	LIMA MINE	0.693		2.5 1330.	697.	896.	0_	0.	78.0	77.9	6.9	6.9	720.		54.00	53.9	0.91	0.9
120875	LIMA MINE	0.904		2.5 1350.	650.		0.	0.		93.7	8.0	10.4				65.1	0.80	1.0
121575	LIMA MINE	2,209		2.7 1190.		1679.	0.	0.		206.7	3.2	10.2		2020.			0.80	2.5
122275	LIMA MINE	1.291		2.6 1230.		1305.	0.	0.		145.0	5.4	10.0		1320.			0.85	1.6
122975	LIMA MINE	2.209		2.8 1000.		1457.	0.	0.		190.8	0.4	1.3		1813.			0.62	2.0
010576	LIMA MINE	4.010		.6 1190.		3314.	0.	0.		415.7	1.9	11.0		3724.			1.70	9.8
011276	LIMA MINE	1.400		.6 1320.		1274.	0.	0.		173.4	5.0	10.1		1562.		92.7	1.30	2.6
011976	LIMA MINE	1.189		.5 1360.		1099.	0.	0.		154.1	2.6	4.5		1336.		82.2	1.20	2.1
012676	LIMA MINE	3.738		.0 665.		1405.	0.	0.		199.2	1.7	9.2		1561.			0.53	2.9
020276	LIMA MINE	1.736		.7 1180.		1180.	0.	0.		175.0	4.9	12.3		1570.			1.00	2.5
020976	LIMA MINE	1.400	2.7	2.7 1230.	537.	1083.	0.	0.	75.0	147.2	1.0	2.0	655.	1280.	44.00	88.7	1.10	2.2

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WEST VIRGINIA ACID MINE DRAINAGE STUDY SNOWY CREEK-LAUREL RUN WATER SAMPLING DATA

LOCATION DOG1 LIMA MINE

	* .	FLOW	Pi	ŧ	SPEC	ACIDITY	ALKALINI	LTY .	IRON(TO	OTALO	IRON (FFRR)	SULF	ATE	AI III	INUM	MANC	ANESE
DATE	IDENT	CMM	FIELD	LAB	COND	MG/L KG/D	MG/L K	G/D		– -	MG/L	KG/D		KG/D	MG/L	KG/D	MG/L	KG/D
C21676 C22376 C3U876 C3U876 C31576 U32276 C32976 C40576 U41276 C41276 C41276 U42676	LIMA MINE	4-914 2-875 1-400 0-826 2-464 4-914 1-189 4-333 1-189 0-765 1-189	2.7 2.9 2.9 2.7 2.7 2.6 2.6 2.6 2.7	2.6 2.6 2.7 2.6 2.8 2.7	750. 980. 1080. 1150. 1040. 960. 1080. 970. 1040. 870. 870.	293. 2073. 406. 1681. 510. 1028. 584. 694. 458. 1625. 334. 2363. 468. 802. 342. 2134. 482. 825. 562. 619. 380. 651.	0.	00000000000	68.0 2 77.0 1 80.0 61.0 2 43.0 3 63.0 1 42.0 2 72.0 1 73.0	281.5 155.2 95.1 216.4 304.2 107.9 262.0	1.4 1.2 4.3 2.4 1.3 1.2 1.5 0.9 6.3 7.2	9.9 5.0 8.7 2.9 4.6 8.5 2.6 5.6 10.8 7.9 2.4	470.	2111. 1260. 889. 2075. 3007. 1028. 3494. 1182. 771.	33.00 42.00 46.00 37.00 31.00 40.00 30.00	136.6 84.7 54.7 131.3 219.3 68.5 187.2	1.00 1.60 1.10 1.30 1.00 0.75 1.20 1.40 1.60 1.20	7.1 6.6 2.2 1.5 3.5 5.3 2.1 11.9 2.4 1.5

AVERAGES FOR 52 SAMPLINGS.

FLOW PH SPEC ACIDITY ALKALINITY IRON(TOTAL) IRON(FERR) SULFATE ALUMINUM MANGANESE CMM FIELD LAB COND MG/L KG/D MG/L

WEST VIRGINIA ACID MINE DRAINAGE STUDY SNOWY CREEK-LAUREL RUN WATER SAMPLING DATA

LOCATION 0002 LIMA MINE

		FLO	W P	н	SPEC	ACII	YTIO	ALKALI	NITY	IRON(T	OTAL)	IRON	FERR)	รบเ	FATE	ALUM	INUM	MANG	SANESE
		CMM	FIELD	LAB			KG/D			MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
DATE	IDEN	Т																	
051475	LIMA M	INE DRY																	
052875	LIMA M	INE DRY																	
061175	LIMA M	INE DRY																	
662475	LIMA M	INE DRY																	
070775	LIMA M	INE DRY																	
072175	LIMA M	INE DRY		•															
080475	LIMA M	INE DRY																	
081875	LIMA M	INE 0.01	9 2.5	2.6	2050.	1300.	35.	0.	0.	310.0	8.3	128.0	3.4	1340.	36.	74.00	2.0	7.90	0.2
082575	LIMA M	INE 0.15	1 2.7	2.6	1020.	562.	122.	0.	0.	37.0	8.1	4.0	0.9	429.		30.00	6.5	2.40	0.5
090275	LIMA M	INE 0.56	7 2.7	2.7	1070.	380.	311.	0.	0.	28.0	22.9	0.1	0.1	410.	335.	24.00	19.6	1.50	1.2
091675	LIMA M	INE DRY																	
C92975	LIMA M	INE DRY																	
101375	LIMA M	INE DRY																	
102775	LIMA M	INE DRY																	
111075	LIMA M	INE DRY																	
112475	LIMA MI	INE DRY																	
120875	LIMA M	INE DRY																	
122275	LIMA M	INE DRY																	
010576	LIMA MI	INE DRY							•										
011976	LIMA M	INE DRY																	
020276	LIMA M	INE DRY																	
021676	LIMA MI	INE 0.47	4 2.9	2.8	1300.	446.	304.	0.	0.	63.0	43.0	0.5	0.3	635.	433.	41.00	28.0	2.80	1.9
030176	LIMA M	INE DRY																	
031576	LIMA M	INE DRY																	
032276	LIMA MI			2.7	1240.	507-	207.	0.	0 -	55.0	22.5	0.4	0.2	620.	253.	47.00	19.2	3.50	1.4
032976	LIMA MI						•		•										
040576	LIMA MI			2.6	1110.	497 -	108.	0.	0 -	52.0	11.3	1.8	0.4	630.	137.	46.00	10.0	3.80	0.8
041276	LIMA MI						. 30	•	•								-		

AVERAGES FOR 28 SAMPLINGS,

FLOW PH SPEC ACIDITY ALKALINITY IRON(TOTAL) IRON(FERR) SULFATE ALUMINUM MANGANESE CMM FIELD LAB COND MG/L KG/D MG/L

WEST VIRGINIA ACID MINE DRAINAGE STUDY SNOWY CREEK-LAUREL RUN WATER SAMPLING DATA

LOCATION 0003 SNOWY CREEK

		FLOW	PH		SPEC	ACII	YTIC	ALKALI	NITY	IRON(1	TOTAL)	-IRON	(FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE
		CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	
DATE	IDENT																		
051475	SNOWY CREEK	24.976	5.7	6.4	78.	3.	108.	0.	0.	0.4	14.0	0.1	3.6	10.	345.	0.10	7.	0 0-	
052875	SNOWY CREEK	25.417	5.8	6.8	90.	4.	146.	Ō.	Ŏ.	0.7	26.0	0.2	6.6	10.	366.	0.25	3.6 9.2	0.05 0.16	1.8 5.9
061175	SNOWY CREEK	49.068	5.5	6.2	75.	8.	565.	0.	O.	0.9	63.6	0.3	19.8	10.	707.	0.29	20.5	0.14	9.9
062475	SNOWY CREEK	18.893	6.2	6.5	94.	6.	163.	0.	0.	1.1	29.9	0.6	15.8	12.	326.	0.38	10.3	0.21	5.7
070775	SNOWY CREEK	45.398	6.5	6.8	103.	10.	654.	0.	0.	1.2	78.4	0.6	39.9	10.	654.	0.51	33.3	0.21	13.7
072175	SNOWY CREEK	49.951	6.6	6.3	110.		1007.	0.	0.	1.0	70.5	0.8	56.1	12.	863.	0.00	0.0	0.00	0.0
080475	SNOWY CREEK	21.612	6.8	6.6	139.	4.	124.	0.	0.	1.0	31.1	0.4	13.1	10.	311.	0.00	0.0	0.00	0.0
081875	SNOWY CREEK	171.091	5.5	6.4	69.		2464.	0.	0.	0.6	150.3	0.3	73.9		2267.	0.00	0.0	0.00	0.0
090275	SNOWY CREEK	283.906	5.6	6.5	63.	1.	409.	0.	0.	0.4	163.5		155.4		4088.	0.00	0.0	0.00	0.0
091675	SNOWY CREEK	40.029	7.0	6.7	94.	5.	288.	0.	0.	1.2	69.2	0.2	13.8	17.		0.00	0.0	0.00	0.0
092975	SNOWY CREEK	54.437	6.4	6.8	82.	0.	0.	5.	392.	0.4	31.4	0.2	15.7	8.	658.	0.00	0.0	0.00	0.0
101375	SNOWY CREEK	61.759	6.3	6.8	80.	Ö.	0.	2.	178.	0.7	62.3	0.2	19.6	4.	329.	0.00	0.0	0.00	0.0
102775	SNOWY CREEK	38.245	6.9	6.2	95.	10.	551.	0.	0.	0.5	27.5	0.3	16.5	10.	551.	0.00	0.0	0.00	0.0
111075	SNOWY CREEK	24.857	6.6	6.6	112.	0.	0.	7.	251.	0.6	21.5	0.4	14.3	9.	322.	0.00	0.0	0.00	0.0
112475	SNOWY CREEK	33.980	6.8	6.9	77.	0.	0.	13.	636.	0.5	22.0	0.2	9.8	9.	431.	0.00	0.0	0.00	0.0
120875	SNOWY CREEK	51.446	6.7	6.9	82.	0.	0.	9.	667.	0.5	37.8	0.2	14.8	11.	815.	0.00	0.0	0.00	0.0
122275	SNOWY CREEK	49.272	6.3	6.7	81.	0.	0.	5.	355.	0.4	27.7	0.2	11.4	10.	710.	0.00	0.0	0.00	0.0
010576	SNOWY CREEK	160.727	5.8	6.4	65.	0.	0.	5.	1157.	0.3	62.5	0.1	11.6	11.	2546.	0.00	0.0	0.00	0.0
020276	SNOWY CREEK	80.449	6.0	6.2	112.	0.	0.	6.	695.	0.5	53.3	0.1	5.8	11.	1274.	0.15	17.4	0.10	11.6
021676	SNOWY CREEK	218.664	5.8	6.1	60.	0.	0.	2.	630.	0.4	113.4	0.1	15.7	10.	3149.	0.00	0.0	0.00	0.0
030176	SNOWY CREEK	52.602	6.5	6.2	73.	1.	76.	0.	0.	0.4	31.8	0.1	3.8	15.	1136.	0.20	15.1	0.08	6.1
031576	SNOWY CREEK	97.524	6.3	6.2	70.	o.	0.	2.	281.	0.4	54.8	0.1	7.0		1545.	0.00	0.0	0.00	0.0
C32976	SNOWY CREEK	49.221	6.8	6.2	74.	0.	0.	3.	213.	0.4	26.9	0.1	3.5	12.	851.	0.00	0.0	0.00	0.0
041276	SNOWY CREEK	40.386	6.0	6.2	75.	0.	0.	6.	349.	0.3	18.0	0.1	2.9	9.	535.	0.00	0.0	0.00	0.0
042676	SNOWY CREEK	49.221	6.5	6.4	91.	0.	0.	9.	638.	0.8	55.3	0.2	11.3	11.	780.	0.00	0.0	0.00	0.0

AVERAGES FOR 25 SAMPLINGS.

 FLOW
 PH
 SPEC
 ACIDITY
 ALKALINITY
 IRON(TOTAL)
 IRON(FERR)
 SULFATE
 ALUMINUM
 MANGANESE

 CMM
 FIELD
 LAB
 COND
 MG/L
 KG/D
 MG/L
 KG/D

WEST VIRGINIA ACID MINE DRAINAGE STUDY SNOWY CREEK-LAUREL RUN WATER SAMPLING DATA

LOCATION 0004 SNOWY CREEK

		FLOW PH	SPEC	ACIDITY	ALKALINITY	IRON(TOTAL)	IRON(FERR)	SULFATE	ALUMINUM	MANGANESE
		CMM FIELD	LAB COND	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D
DATE	IDENT									
051475	SNOWY CREEK	57.580 5.0	6.4 79.	22. 1824.	0. 0.	1.2 99.5	0.3 21.6	19. 1575.	0.30 24.9	0.05 4.1
052175	SNOWY CREEK	54.454 5.3	6.5 83.	28. 2196.	0. 0.	1.2 94.1	0.2 18.8	16. 1255.	0.81 63.5	0.06 4.7
052875	SNOWY CREEK	46,553 5.4	6.7 93.	29. 1944.	0. 0.	1.3 87.1	0.3 21.5	20. 1341.	0.95 63.7	0.14 9.4
(61175	SNOWY CREEK	52.619 5.5	6.3 94.	35. 2652.	0. 0.	1.4 106.1	0.4 33.3	18. 1364.	0.79 59.9	0.11 8.3
062475	SNOWY CREEK	20.558 5.6	6.4 101.	30. 888.	0. 0.	1.4 41.4	0.4 10.7	20. 592.	1.00 29.6	0.16 4.7
676775	SNOWY CREEK	GRAB SMPL 6.3	6.8 105.	32.	0.	1.4	0.4	16.	0.95	0.15
072175	SNOWY CREEK	GRAB SMPL 6.6	6.7 137.	24.	0.	1.4	0.5	16.	0.00	0.00
080475	SNOWY CREEK	GRAB SMPL 6.4	6.6 132.	10.	0.	1.2	0.4	17.	0.00	0.00
081875	SNOWY CREEK	GRAB SMPL 5.3	6.3 86.	31.	0.	1.5	0.7	18.	0.00	0.00
090275	SNOWY CREEK	GRAB SMPL 5.2	6.3 68.	21.	0.	1.1	0.4	16.	0.00	0.00
091675	SNOWY CREEK	GRAB SMPL 5.8	6.7 94.	2.	0.	0.6	0.3	9.	0.00	0.00
101375	SNOWY CREEK	GRAB SMPL 5.7	6.7 84.	5.	0.	1.2	0.5	20.	0.00	0.00
111075	SNOWY CREEK	GRAB SMPL 5.7	6.6 106.	6.	0.	1.5	0.1	18.	0.00	0.00
120875	SNOWY CREEK	GRAB SMPL 5.3	6.5 86.	5.	0.	1.4	0.1	20.	0.00	0.00
020276	SNOWY CREEK	GRAB SMPL 5.6	5.9 114.	7.	0.	1.3	0.1	20.	0.75	0.10
021676	SNOWY CREEK	GRAB SMPL 4.8	6.0 68.	5.	0.	1.0	0.1	17.	0.00	0.00
030176	SNOWY CREEK	GRAB SMPL 5.2	5.9 82.	10.	0.	1.9	0.2	23.	0.00	0.00
641276	SNOWY CREEK	GRAB SMPL 5.4	6.1 83.	3.	0.	1.4	0.3	20.	0.00	0.00
042676	SNOWY CREEK	GRAB SMPL 6.2	6.1 98.	1.	o.	1.3	0.3	19.	0.00	0.00

AVERAGES FOR 19 SAMPLINGS.

FLOW PH SPEC ACIDITY ALKALINITY IRON(TOTAL) IRON(FERR) SULFATE ALUMINUM MANGANESE CMM FIELD LAB COND MG/L KG/D MG/L

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WEST VIRGINIA ACID MINE DRAINAGE STUDY SNOWY CREEK-LAUREL RUN WATER SAMPLING DATA

LOCATION DODS SNOWY CREEK

		FLOW	PH	1	SPEC	ACIDITY	ALKALI	NITY	IRON (TO	TAL)	IRON(FERR)	SULFATE	ALUM	INUM	MANG	ANESE
		CMM	FIELD	LAB	COND	MG/L KG/D	MG/L	KG/D	MG/L	KG/D	MG/L KG/D		MG/L		MG/L	
DATE	1 DENT												,			
651475	SNOWY CREEK	60.043	5.0	6.4	81.	22. 1902.	0.	0.		78.7	0.1 12.1	19. 1643.	0.20	17.3	0.10	8_6
052875	SNOWY CREEK	47.369	5.4	6.7	90.	24. 1637.	0.	0.		75.0	0.3 20.5	20. 1364.	0.40	27.3	0.15	10.2
C61175	SNOWY CREEK	53.689	5.5	6.4	82.	35. 2706.	0.	0.		77.3	0.3 20.1	18. 1392.	0.55	42.5	0.10	7.7
062475	SNOWY CREEK	20.592	5.7	6.6	106.	18. 534.	0.	0.	1.4	41.5	0.3 10.1	23. 682.	0.75	22.2	0.15	4.4
C70775	SNOWY CREEK	46.162	6.3	6.8	112.	18. 1197.	0.	0.		93.1	0.4 26.6	19. 1263.	0.80	53.2	0.15	10.0
072175	SNOWY CREEK	59.296	6.5	6.7	120.	14. 1195.	0.	0.		28.1	0.4 34.2	16. 1366.	0.00	0.0	0.00	0.0
080475	SNOWY CREEK	22.835	6.5	6.7	133.	12. 395.	0.	0.	1.3	42.7	0.6 18.4	18. 592.	0.00	0.0	0.00	0.0
081875	SNOWY CREEK	203.203	5.4	6.2	87.	32. 9364.	0.	0.	1.6 4	-	0.8 234.1	18. 5267.	0.00	0.0	0.00	0.0
090275	SNOWY CREEK	292.062	5.3	6.3	68.	22. 9253.	0.	0.	1.2 5	04.7	0.5 210.3	15. 6309.	0.00	0.0	0.00	0.0
D91675	SNOWY CREEK	44.888	5.8	6.7	96.	11. 711.	0.	0.		71.1	0.2 14.2	18. 1163.	0.00	0.0	0.00	0.0
092975	SNOWY CREEK	54.810	5.5	6.6	84.	18. 1421.	0.	0.	1.2	94.7	0.4 31.6	18. 1421.	0.00	0.0	0.00	0.0
101375	SNOWY CREEK	67.893	5.7	6.7	88.	9. 880.	0.	0.	1.1 1	07.5	0.5 48.9	20. 1955.	0.00	0.0	0.00	0.0
111075	SNOWY CREEK	27.541	5.8	6.6	100.	5. 198.	0.	0.	1.1	43.6	0.1 4.0	20. 793.	0.00	0.0	0.00	0.0
112475	SNOWY CREEK	39.230	5.2	6.6	89.	9. 508.	0.	0.	1.4	79.1	0.4 22.6	20. 1130.	0.00	0.0	0.00	0.0
120875	SNOWY CREEK	57.308	5.5	6.6	86.	5. 413.	0.	0.	1.2	99.0	0.4 36.3	23. 1898.	0.00	0.0	0.00	0.0
122275	SNOWY CREEK	65.429	5.3	6.4	92.	13. 1225.	0.	0.	1.4 1	31.9	0.4 37.7	24. 2261.	0.00	0.0	0.00	0.0
010576	SNOWY CREEK	205.072	4.9	5.7	75.	9. 2658.	0.	0.	1.3 3	83.9	0.2 59.1	21. 6201.	0.00	0.0	0.00	0.0
011976	SNOWY CREEK	55.779	4.8	5.0	93.	25. 2008.	0.	0.	1.8 1	44.6	0.2 12.9	25. 2008.	0.00	0.0	0.00	0.0
020276	SNOWY CREEK	97.609	5.4	6.0	85.	12. 1687.	0.	0.	1.4 1	96.8	0.1 14.1	22. 3092.	0.55	77.3	0.10	14.1
021676	SNOWY CREEK	303.955	5.3	6.1	66.	5. 2188.	0-	0.	1.0 4	37.7	0.1 43.8	18. 7879.	0.00	0.0	0.00	0.0
030176	SNOWY CREEK	72.548	5.0	6.3	82.	11. 1149.	0.	0.	1.9 1	98.5	0.1 10.4	16. 1672.	0.80	83.6	0.14	14.6
031576	SNOWY CREEK	125.218	5.0	5.9	76.	10. 1803.	Õ.	0.	1.2 2	16.4	0.2 43.3	20. 3606.	0.00	0.0	0.00	0.0
032976	SNOWY CREEK	67.145	5.5	6.3	82.	8. 774.	Ö.	Ō.	1.1 1		0.3 27.1	18. 1740.	0.00	0.0	0.00	0.0
041276	SNOWY CREEK	57.308	5.4	6.4	88.	8. 660.	Ö.	Ö.		99.0	0.2 14.9	21. 1733.	0.00	0.0	0.00	0.0
042676	SNOWY CREEK	83.031	6.0	6.4	98.	9. 1076.	ŏ.	o.		67.4	0.3 38.3	19. 2272.	0.00	0.0	0.00	0.0

AVERAGES FOR 25 SAMPLINGS.

FLOW	PH	1	SPEC	ACIDITY	ALKAL	INITY	IRON(TOTAL)	IRON	(FERR)	SUL	FATE	ALUF	IINUM	MANG	ANESE
CMM	FIELD	LAB	COND	MG/L KG	D MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
				45 400		_						2420				
89.201	5.5	6.4	90.	15. 190	. O.	0.	1.3	165.5	0.3	41.8	19.	2428.	U.36	40.2	U.U8	10.0

LOCATION 0006 SNOWY CREEK

		FLOW	PI	i	SPEC	ACID	ITY	ALKALI	NITY	IRON(T	OTAL)	IRON (FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE
		CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L		MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L		MG/L	KG/D
DATE	IDENT																		
051475	SNOWY CREEK	2.895	5.3	6.3	30.	18.	75.	0.	0.	1.1	4.6	0.2	1.0	8.	34.	0.10	0.4	0.05	0.2
052875	SNOWY CREEK	3.160	5.5	6.9	32.	33.	150.	o.	o.	1 - 4	6.4	0.5	2.4	10.	46.	0.10	0.5	0.13	0.6
061175	SNOWY CREEK	0.946	5.2	6.6	32.	37.	50.	0.	0.	1.2	1.6	0.4	0.5	8.	11.	0.20	0.3	0.10	0.1
062475	SNOWY CREEK	0.095	5.8	6.7	36.	36.	5.	0.	o.	1.4	0.2	0.3	0.0	10.	1.	0.21	0.0	0.19	0.0
070775	SNOWY CREEK	0.189	5.5	6.8	37.	30.	8 .	ő.	o.	1.4	0.4	0.4	0.1	9.	2.	0.40	0.1	0.59	0.2
072175	SNOWY CREEK	0.095	5.6	6.6	45.	36.	5.	ő.	ő.	2.4	0.3	0.6	0.1	7.	1.	0.00	0.0	0.00	0.0
080475	SNOWY CREEK	0.037	5.6	6.3	40.	20.	í.	o.	ő.	0.4	0.0	0.1	0.0	5.	o.	0.00	0.0	0.00	0.0
081875	SNOWY CREEK	1.135	5.1	5.7	37.	20.	33.	Ö.	o.	0.9	1.4	0.4	0.7	9.	15.	0.00	0.0	0.00	0.0
090275	SNOWY CREEK	1.886	5.3	5.7	30.	17.	46.	o.	ő.	0.7	2.0	0.4	1.2	6.	17.	0.00	0.0	0.00	0.0
091675	SNOWY CREEK	0.758	5.5	6.8	40.	11.	12.	õ.	Ö.	1.2	1.3	0.2	0.3	6.	7.	0.00	0.0	0.00	0.0
092975	SNOWY CREEK	0.758	5.3	6.5	33.	10.	11.	Õ.	Ö.	0.9	1.0	0.4	0.4	5.	5.	0.00	0.0	0.00	0.0
101375	SNOWY CREEK	1.325	5.5	6.7	30.	8.	15.	ō.	Ö.	1.2	2.3	0.5	1.0	5.	10.	0.00	0.0	0.00	0.0
102775	SNOWY CREEK	0.567	5.7	6.5	38.	14.	11.	ő.	ō.	1.0	0.8	0.4	0.3	5.	4.	0.00	0.0	0.00	0.0
111075	SNOWY CREEK	0.758	5.7	6.6	38.	11.	12.	0.	õ.	1.6	1.7	0.4	0.4	5.	5.	0.00	0.0	0.00	0.0
112475	SNOWY CREEK	0.379	5.4	6.5	44.	11.	6.	o.	ő.	1.9	1.0	0.6	0.3	8.	4.	0.00	0.0	0.00	0.0
120875	SNOWY CREEK	0.567	5.6	6.5	42.	11.	9.	ō.	ő.	1.7	1.4	0.5	0.4	8.	7.	0.00	0.0	0.00	0.0
122275	SNOWY CREEK	0.758	5.4	6.6	32.	7.	8.	Ö.	o.	0.6	0.6	0.2	0.2	6.	7.	0.00	0.0	0.00	0.0
010576	SNOWY CREEK	2.650	5.2	6.2	32.	5.	19.	ŏ.	o.	0.3	1.1	0.1	0.2	5.	20.	0.00	0.0	0.00	0.0
011976	SNOWY CREEK	FROZE	EN																
020276	SNOWY CREEK	FROZE	EN																
021676	SNOWY CREEK	4.543	5.2	6.1	30.	0.	0.	1.	7.	0.3	1.8	0.1	0.3	3.	18.	0.00	0.0	0.00	0.0
030176	SNOWY CREEK	0.637	5.3	6.3	32.	11.	10.	0.	0.	0.8	0.8	0.3	0.3	7.	6.	0.00	0.0	0.00	0.0
031576	SNOWY CREEK	0.637	5.3	6.1	34.	1.	1.	ō.	0.	0.4	0.4	0.3	0.2	7.	6.	0.00	0.0	0.00	0.0
032976	SNOWY CREEK	0.552	5.5	6.3	33.	10.	8.	ŏ.	ō.	0.9	0.7	0.4	0.3	8.	6.	0.00	0.0	0.00	0.0
041276	SNOWY CREEK	0.474	5.3	6.2	33.	5.	3.	ō.	Ö.	0.9	0.6	0.4	0.3	6.	4.	0.00	0.0	0.00	0.0
042676	SNOWY CREEK	0.379	5.6	6.0	47.	10.	5.	o.	o.	2.6	1.4	0.8	0.4	9.	5.	0.00	0.0	0.00	0.0

AVERAGES FOR 26 SAMPLINGS.

LOCATION 0008 SNOWY CREEK

		FLOW			SPEC	ACID		ALKALI		IRONCT		IRON (FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE
		CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L			KG/D
DATE	IDENT										*1								
051475	SNOWY CREEK	1.261	5.4	6.6	25.	4.	7.	0.	0.	0.2	0.3	0.1	0.2	4	44				
052875	SNOWY CREEK	6.150	5.0	6.9	26.	8.	71.	õ.	ő.	0.2	1.6	0.1	0.4	6. 8.	11. 69.	0.10	0.2	0.03	0.1
061175	SNOWY CREEK	0.663	5.0	6.5	25.	17.	16.	ō.	Ö.	0.1	0.1	0.1	0.0	6.	6.	0.10	1.9	0.01	0.1
062475	SNOWY CREEK	0.379	5.5	6.5	25.	25.	14.	ō.	0.	0.2	0.1	0.1	0.0	7.	4.	0.20	0.1	0.05	0.0
070775	SNOWY CREEK	0.189	5.5	6.7	29.	8.	2.	Ö.	Ö.	0.2	0.1	0.1	0.0	6.	Ž.	0.15	0.0		0.0
072175	SNOWY CREEK	0.379		6.6	46.	20.	11.	ō.	ō.	0.6	0.3	0.1	0.0	12.	7.	0.00	0.0	0.02	0.0
080475	SNOWY CREEK	0.095	5.5	6.2	32.	22.	3.	ō.	ő.	0.4	0.1	0.1	0.0	7.	j.	0.00	0.0	0.00	0.0
081875	SNOWY CREEK	0.454	5.0	5.7	31.	23.	15.	Ō.	Õ.	0.4	0.3	0.2	0.1	8.	5.	0.00	0.0	0.00	0.0 0.0
090275	SNOWY CREEK	1.211	5.0	5.7	26.	15.	26.	ō.	0.	0.2	0.3	0.1	0.1	6.	10.	0.00	0.0	0.00	0.0
091675	SNOWY CREEK	0.189	5.2	6.5	29.	9.	2.	0.	Ö.	0.2	0.0	0.1	0.0	6.	ž.	0.00	0.0	0.00	0.0
092975	SNOWY CREEK	0.567	4.9	6.2	28.	5.	4.	0.	0.	0.1	0.1	0.1	0.1	6.	5.	0.00	0.0	0.00	0.0
101375	SNOWY CREEK	0.189	5.0	6.0	30.	7.	2.	0.	0.	0.2	0.1	0.1	0.0	6.	ź.	0.00	0.0	0.00	0.0
102775	SNOWY CREEK	0.663	5.3	6.4	29.	4.	4.	0.	0.	0.2	0.2	0.1	0.1	6.	6.	0.00	0.0	0.00	0.0
111075	SNOWY CREEK	0.284	5.3	6.6	32.	0.	0.	0.	O.	0.4	0.2	0.1	0.0	6.	ž.	0.00	0.0	0.00	0.0
112475	SNOWY CREEK	0.567	5.4	6.3	28.	1.	1.	ō.	ō.	0.2	0.2	0.1	0.0	6.	5.	0.00	0.0	0.00	0.0
120875	SNOWY CREEK	0.379	5.4	6.4	31.	2.	1.	Ö.	Ö.	0.3	0.2	0.1	0.0	3.	2.	0.00	0.0	0.00	0.0
122275	SNOWY CREEK	0.284	4.9	6.3	28.	4.	2.	o.	Ö.	0.5	0.2	0.1	0.0	9.	4.	0.00	0.0	0.00	0.0
010576	SNOWY CREEK	0.474		5.7	39.	3.	2.	ō.	Ō.	0.1	0.1	0.1	0.0	6.	4.	0.00	0.0	0.00	0.0
011976	SNOWY CREEK	FROZ			- , •			•	••	•••		•••		••	•	0.00	0.0	0.00	0.0
020276	SNOWY CREEK	0.561	5.2	6.4	30.	3.	2.	0.	0.	0.2	0.2	0.1	0.0	6.	5.	0.00	0.0	0.00	0.0
021676	SNOWY CREEK	1.893	4.3	6.4	32.	5.	14.	ō.	o.	0.1	0.3	0.1	0.1	6.	16.	0.00	0.0	0.00	0.0
030176	SNOWY CREEK	0.474	4.7	6.2	29.	4.	3.	o.	ŏ.	0.1	0.1	0.1	0.0	6.	4.	0.00	0.0	0.00	0.0
031576	SNOWY CREEK	0.663	4.8	5.9	27.	5.	š.	o.	ő.	0.1	0.1	0.1	0.0	5.	5.	0.00	0.0	0.00	0.0
032976	SNOWY CREEK	0.284	4.9	6.3	28.	6.	ž.	o.	ŏ.	0.2	0.1	0.1	0.0	6.	ź.	0.00	0.0	0.00	0.0
041276	SNOWY CREEK	0.284	4.8	6.0	29.	7.	3.	ŏ.	ŏ.	0.2	0.1	0.1	0.0	5.	2.	0.00	0.0	0.00	0.0
042676	SNOWY CREEK	0.379	5.2	6.2	31.	7.	4.	ŏ.	Ö.	0.2	0.1	0.1	0.0	6.	3.	0.00	0.0	0.00	0.0
												- • •							

AVERAGES FOR 26 SAMPLINGS,

F	LOW	PH		SPEC	ACID	ITY	ALKALI	NITY	IRON (T	OTAL)	IRON (FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE
C	MM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/Ł	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
0.	727	5.1	6.3	30.	8.	8.	0.	0.	0.2	0.2	0.1	0.1	7.	7.	0.44	0.5	0.04	0.0

LOCATION 0009 SNOWY CREEK

		FLOW	PH	SPEC	ACIDITY	ALKALINITY	IRON(TOTAL)	IRON(FERR)	SULFATE	ALUMINUM	MANGANESE
		CMM	FIELD LA		MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D
DATE	IDENT	2		• • • • • • • • • • • • • • • • • • • •							
051475	SNOWY CREEK	75.810	5.2 6.	2 76.	18. 1965.	0. 0.	0.6 69.9	0.1 10.9	16. 1747.	0.10 10.9	0.06 6.6
052875	SNOWY CREEK	60.893	5.9 6.	9 84.	22. 1929.	0. 0.	1.0 87.7	0.4 31.6	17. 1491.	0.35 30.7	0.14 12.3
061175	SNOWY CREEK	65 - 616	5.6 6.	83.	32. 3024.	0. 0.	0.9 86.9	0.3 30.2	17. 1606.	0.21 19.8	0.12 11.3
062475	SNOWY CREEK	28.085	5.7 6.	6 99.	24. 971.	0. 0.	1.3 52.6	0.2 8.9	19. 768.	0.32 12.9	0.14 5.7
070775	SNOWY CREEK	56.000	6.6 6.	3 109.	29. 2339.	0. 0.	1.3 104.8	0.3 25.0	15. 1210.	0.85 68.5	0.11 8.9
072175	SNOWY CREEK	67.468	6.7 6.	6 120.	16. 1554.	0. 0.	1.2 116.6	0.3 29.1	19. 1846.	0.00 0.0	0.00 0.0
080475	SNOWY CREEK	23.786	6.9 6.	8 113.	20. 685.	0. 0.	0.9 31.2	0.1 4.1	15. 514.	0.00 0.0	0.00 0.0
081875	SNOWY CREEK	205.581	5.5 6.	1 95.	44.13026.	0. 0.	1.5 444.1	0.5 142.1	18. 5329.	0.00 0.0	0.00 0.0
090275	SNOWY CREEK	297.328	5.3 6.	3 65.	30.12845.	0. 0.	1.1 471.0	0.4 162.7	15. 6422.	0.00 0.0	0.00 0.0
091675	SNOWY CREEK	49.458	6.1 6.	94.	20. 1424.	0. 0.	1.0 71.2	0.3 19.9	16. 1140.	0.00 0.0	0.00 0.0
092975	SNOWY CREEK	56.136	5.7 6.	7 81.	10. 808.	0. 0.	0.7 56.6	0.2 16.2	17. 1374.	0.00 0.0	0.00 0.0
101375	SNOWY CREEK	75.335	5.8 6.	86.	5. 542.	0. 0.	0.8 85.7	0.3 32.5	20. 2170.	0.00 0.0	0.00 0.0
102775	SNOWY CREEK	53.145	6.4 6.	7 88.	6. 459.	0. 0.	0.7 53.6	0.3 23.0	19. 1454.	0.00 0.0	0.00 0.0
111075	SNOWY CREEK	30.311	6.5 6.	3 107.	19. 829.	0. 0.	1.3 56.7	0.5 21.8	17. 742.	0.00 0.0	0.00 0.0
112475	SNOWY CREEK	40.182	6.0 6.		14. 810.	0. 0.	0.9 52.1	0.4 20.8	19. 1099.	0.00 0.0	0.00 0.0
120875	SNOWY CREEK	58.259	5.9 6.	85.	10. 839.	0. 0.	0.9 73.8	0.1 11.7	18. 1510.	0.00 0.0	0.00 0.0
122275	SNOWY CREEK	66.534	5.6 6.	132.	24. 2299.	0. 0.	1.1 105.4	0.2 23.0	20. 1916.	0.00 0.0	0.00 0.0
010576	SNOWY CREEK	204.902	5.0 5.		16. 4721.	0. 0.	1.2 354.1	0.2 47.2	20. 5901.	0.00 0.0	0.00 0.0
011976	SNOWY CREEK	60.961	5.4 5.	86.	9. 790.	0. 0.	0.6 48.3	0.1 4.4	15. 1317.	0.00 0.0	0.00 0.0
020276	SNOWY CREEK	53.145	5.6 6.	2 94.	8. 612.	0. 0.	0.8 62.0	0.1 7.7	18. 1378.	0.00 0.0	0.00 0.0
021676	SNOWY CREEK	260.800	5.1 6.	1 64.	2. 751.	0. 0.	0.7 266.6	0.1 37.6	16. 6009.	0.00 0.0	0.00 0.0
030176	SNOWY CREEK	67.961	5.5 6.		7. 685.	0. 0.	0.8 78.3	0.2 21.5	19. 1859.	0.00 0.0	0.00 0.0
031576	SNOWY CREEK	109.706	5.3 6.		8. 1264.	0. 0.	1.0 158.0	0.2 25.3	17. 2686.	0.00 0.0	0.00 0.0
032976	SNOWY CREEK	60.961	5.8 6.		2. 176.	0. 0.	0.5 46.5	0.1 4.4	16. 1405.	0.00 0.0	0.00 0.0
041276	SNOWY CREEK	48.252	5.6 6.		4. 278.	0. 0.	0.6 43.1	0.1 3.5	19. 1320.	0.00 0.0	0.00 0.0
042676	SNOWY CREEK	75.403	6.6 6.	_	3. 326.	0. 0.	1.3 141.2	0.2 23.9	19. 2063.	0.00 0.0	0.00 0.0

AVERAGES FOR 26 SAMPLINGS,

LOCATION 0010 SNOWY CREEK

		FLOW	PH		SPEC	ACIDITY	ALKALI		IRON(TOTAL)	IRON(FERR)	SULFATE	ALUMINUM	MANGANESE
DATE	IDENT	CMM	FIELD	LAB	COND	MG/L KG/D	MG/L	KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D
052875	SNOWY CREEK	75.997	4.9	5 - 6	94.	37. 4049.	0.	0.	2.3 251.7	1.1 120.4	36. 3940.	1.20 131.3	0.20 21.9
061175	SNOWY CREEK	83.337	5.0	6.0	88.	51. 6120.	0.	0.	1.8 216.0	1.2 144.0	30. 3600.	1.10 132.0	0.16 19.2
062475	SNOWY CREEK	35.951	5.4	5.9	106.	41. 2123.	0.	0.	1.9 98.4	0.5 23.8	36. 1864.	1.10 56.9	0.25 12.9
070775	SNOWY CREEK	61.369	5.5	6.4	120.	38. 3358.	0.	0.	1.7 150.2	0.6 53.0	38. 3358.	1.20 106.0	0.20 17.7
072175	SNOWY CREEK	74.723	6.0	6.6	126.	22. 2367.	0.	0.	2.0 215.2	0.7 75.3	35. 3766.	0.00 0.0	0.00 0.0
080475	SNOWY CREEK	27.456	5.8	6.4	140.	30. 1186.	0.	0.	2.2 87.0	0.2 8.7	47. 1858.	0.00 0.0	0.00 0.0
081875	SNOWY CREEK	295.629	5.2	6.0	88.	30.12771.	0.	0.	1.5 638.6	0.9 391.6	20. 8514.	0.00 0.0	0.00 0.0
090275	SNOWY CREEK	467.910	4.7	5.4	65.	38.25604.	0.	ο.	1.3 875.9	0.2 107.8	19.12802.	0.00 0.0	0.00 0.0
091675	SNOWY CREEK	66.058	4.8	5.6	150.	32. 3044.	0.	0.	2.5 237.8	1.4 133.2	34. 3234.	1.10 104.6	0.19 18.1
092975	SNOWY CREEK	82.997	4.7	5.0	97.	35. 4183.	0.	0.	2.4 286.8	2.0 239.0	44. 5259.	0.00 0.0	0.00 0.0
101375	SNOWY CREEK	94.483	5.3	6.3	93.	28. 3810.	0.	0.	1.9 258.5	1.9 258.5	32. 4354.	0.00 0.0	0.00 0.0
102775	SNOWY CREEK	76.660	5.0	5.1	99.	36. 3974.	0.	0.	2.3 253.9	1.4 154.5	35. 3864.	0.00 0.0	0.00 0.0
111075	SNOWY CREEK	44.225	4.8	5.7	119.	28. 1783.	0.	0.	2.6 165.6	1.6 101.9	44. 2802.	0.00 0.0	0.00 0.0
111775	SNOWY CREEK	80.958	5.2	6.1	90.	15. 1749.	0.	0.	2.1 244.8	1.3 151.6	27. 3148.	0.90 104.9	0.21 24.5
112475	SNOWY CREEK	66.058	5.0	5.4	94.	21. 1998.	0.	Q.	2.5 237.8	2.2 209.3	33. 3139.	0.00 0.0	0.00 0.0
120175	SNOWY CREEK	62.439	5.3	5.6	89.	20. 1798.	0.	0.	2.3 206.8	1.8 161.8	33. 2967.	1.00 89.9	0.19 17.1
120875	SNOWY CREEK	86.208	5.2	6.0	89.	11. 1366.	0.	0.	2.0 248.3	1.6 198.6	26. 3228.	0.00 0.0	0.00 0.0
121575	SNOWY CREEK	139.897	5.4	4.9	82.	32. 6446.	0.	0.	1.9 382.8	1.2 241.7	28. 5641.	0.00 0.0	0.00 0.0
122275	SNOWY CREEK	98.781	5.6	4.7	138.	35. 4979.	0.	0.	2.8 398.3	2.0 284.5	35. 4979.	0.00 0.0	0.00 0.0
122975	SNOWY CREEK	136.737	5.2	5.0	82.	32. 6301.	0.	0.	1.9 374.1	1.3 256.0	27. 5316.	0.00 0.0	0.00 0.0
010576	SNOWY CREEK	309.731	4.0	4.8	91.	36.16056.	0.	0.	2.71204.2	2.1 936.6	31.13826.	0.00 0.0	0.00 0.0
011276	SNOWY CREEK	91.288	4.4	4.8	110.	37. 4864.	0.	0.	2.7 354.9	2.5 328.6	36. 4732.	0.00 0.0	0.00 0.0
011976	SNOWY CREEK	115.771	4.0	4.2	135.	44. 7335.	0.	0.	4.1 683.5	4.1 683.5	45. 7502.	1.50 250.1	0.21 35.0
012676	SNOWY CREEK	207.790	5.0	5.8	98.	30. 8977.	0.	0.	2.5 748.0	1.1 329.1	29. 8677.	0.00 0.0	0.00 0.0
020276	SNOWY CREEK	81.332	4-0	4.6	98.	36. 4216.	o.	0.	2.5 292.8	1.1 128.8	29. 3396.	0.00 0.0	0.00 0.0
020976	SNOWY CREEK	101.941	4.5	4.9	110.	32. 4697.	0.	0.	2.4 352.3	1.9 278.9	29. 4257.	0.00 0.0	0.00 0.0
021676	SNOWY CREEK	469.779	3.6	4.7	76.	10. 6765.	Q.	0.	1.3 879.4	0.9 581.8	22.14883.	0.00 0.0	0.00 0.0
022376	SNOWY CREEK	223.931	4.5	4.3	82.	16. 5159.	0.	0.	1.6 515.9	1.0 322.5	26. 8384.	0.00 0.0	0.00 0.0
030176	SNOWY CREEK	73.296	4.0	4.5	102.	26. 2744.	٥.	0.	2.6 274.4	2.1 221.6	36. 3800.	0.00 0.0	0.00 0.0
630876	SNOWY CREEK	28.781	4.9	4.1	130.	26. 1078.	0.	0.	2.8 116.0	2.5 103.6	41. 1699.	0.00 0.0	0.00 0.0
031576	SNOWY CREEK	163.208	4.0	5.2	79.	17. 3995.	0.	0.	1.8 423.0	1.1 258.5	25. 5875.	0.00 0.0	0.00 0.0
032276	SNOWY CREEK	377.862	4.4	4.7	74.	6. 3265.	0.	0.	1.5 816.2	0.9 500.6	17. 9250.	0.00 0.0	0.00 0.0
032976	SNOWY CREEK	61.505	4.2	4.6	98.	20. 1771.	0.	0.	2.3 203.7	2.2 194.8	35. 3100.	0.00 0.0	0.00 0.0
040576	SNOWY CREEK	219.853	4.0	4.5	75.	8. 2533.	0.	0.	1.4 443.2	0.9 272.3	18. 5699.	0.00 0.0	0.00 0.0
041276	SNOWY CREEK	48.813	4.0	4.7	103.	21. 1476.	0.	0.	2.7 189.8	2.2 154.6	35. 2460.	0.00 0.0	0.00 0.0
041976	SNOWY CREEK	19.284	4 . D	4.3	118.	23. 639.	0.	0.	2.4 66.6	1.6 44.4	40. 1111.	1.20 33.3	0.20 5.6
042676	SNOWY CREEK	88.723	4.0	4.8	119.	14. 1789.	0.	0.	2.1 268.3	1.7 217.2	26. 3322.	0.00 0.0	0.00 0.0
050376	SNOWY CREEK	32.451	5.0	4.7	107.	18. 841.	0.	0.	2.3 107.5	1.4 65.4	32. 1495.	0.00 0.0	0.00 0.0

AVERAGES FOR 38 SAMPLINGS.

LOCATION 0011 SLIME HOLE

		FLOW	PH		SPEC) T Y	ALKALI				IRON (FERR)		FATE		AINUM		ANESE
DATE	IDENT	CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
	102.01																	
051475	SLIME HOLE	1.988	2.7	3.0	1060.	335.	959.	0.	0.	93.0	266.2	90.0 257.6		1703.		83.0	2.00	5.7
052175	SLIME HOLE	1.886	2.8	2.9	1120.	365.	991.	0.				98.0 266.1		1711.		81.5	2.00	5.4
052875	SLIME HOLE	1.886	3.1		1040.	342.	929.	0.	0.		225.4	51.0 138.5		2037.		73.3	1.60	4.3
060475	SLIME HOLE	1.514	2.8		1110.	393.		0.	0.		185.3	79.0 172.2		1275.		56.7	1.80	3.9
061175	SLIME HOLE	2.100	2.5		1120.		1040.	0.	0.		269.1	90.0 272.2		1588.		84.7	1.90	5.7
061675	SLIME HOLE	1.986	2.6		1100.		984.	o.	0.		277.4	90.0 257.4		1573.		80.1	1.90	5.4
062475	SLIME HOLE	2.153	2.5		1160.		1066.	0.	0.		282.1	14.0 43.4		1581.		89.9	2.10	6.5
063075	SLIME HOLE	1.583	2.8		1180.	340.	775.	0.	0.		207.5	91.0 207.5		1140.		66.1	2.00	4.6
070775	SLIME HOLE	1.750	2.8		1180.	330.	832.	0.	o.	_	231.8	30.0 75.6		1323.		73.1	2-00	5.0
071475	SLIME HOLE	1.505	2.5		1160.	354.	767.	0.	0.		190.8	86.0 186.4		1192.		62.9	1.80	3.9
072175	SLIME HOLE	1.325	2.6		1300.	336.	641.	0.	0.		175.6	65.0 124.0		1126.		53.4	1.80	3.4
072875	SLIME HOLE	1.147	2.7		1100.	350.	578.	0.	٥.		148-6	90.0 148.6		974.		44.6	1.80	3.0
080475	SLIME HOLE	1.210	2.7		1200.	356.	620.	0.	0.		165.5	82.0 142.8		1286.		48.8	2.00	3.5
081175	SLIME HOLE	1.084	2.7		1210.	390.	609.	0.	0.		149.8	97.0 151.4		1194.		45.3	2.10	3.3
081875	SLIME HOLE	1.704	2.5		1210.	376.		0.	0.		215.9	15.0 36.8		1423.		68.7	1-90	4.7
082575	SLIME HOLE	2.090	2.7		1120.		1107.	√0.	0.			95.0 285.9		1679.		81.3	2.00	6.0
090275	SLIME HOLE	2.735	2.8		1160.		1871.	0.				134.0 527.8			30.00		2.40	9.5
090875	SLIME HOLE	2.735	2.6		1280.		1650.	0.				108.0 425.4			31.00		2.40	9.5
091675	SLIME HOLE	2.273	2.6		1190.		1309.	0.				75.0 245.5		2046.		98.2	1-90	6.5
092275	SLIME HOLE	2.865	2.7		1060.		1625.	0.				100.0 412.5			30.00		2.00	8.2
092975	SLIME HOLE	2.469	2.9		1280.		1639.	0.		125.0					31.00		2.30	8.2
100675	SLIME HOLE	2.338	2.7		1190.		1397.	0.				115.0 387.1			29.00		1.90	6.4
101375	SLIME HOLE	2.338	2.9		1270.		1414.	o.				43.0 144.8			30.00		2.00	6.7
102075	SLIME HOLE	2.338	2.9		1330.		1488.	0.				118.0 397.2			31.00		2.10	7.1
102775	SLIME HOLE	2.735	2.7		1290.		1698.	0.		120.0					29.00		2.00	7.9 7.1
110375	SLIME HOLE	2.469	2.8		1180.		1600.	0.		115.0		70.0 248.8			29.00		2.00	
111075 111775	SLIME HOLE	2.338	2.7		1060.		1397.	0.		110.0		60.0 202.0		2003.		94.3 89.1	2.00 1.90	6.7 6.0
112475	SLIME HOLE	2.209	2.7		1050.	_	1234.	0.		105.0		60.0 190.8		1902. 1986.		94.3	2.00	6.7
120175	SLIME HOLE	2.338 2.338	2.7 2.8		1090. 1030.		1380. 1387.	0.		110.0		54.0 181.8 100.0 336.7		2087.		90.9	2.00	6.7
120875	SLIME HOLE	2.338	2.6		1180.	-	1448.	0.				120.0 404.0		2138.		90.9	2.10	7.1
121575	SLIME HOLE	2.600	2.8			411.		0. 0.				120.0 449.2			29.00		2.20	8.2
122275	SLIME HOLE	2.739	2.6		1120.		1948.	0.				100.0 394.4			29.00		2.30	9.1
122975	SLIME HOLE	2.600	2.7		980.		1535.					110.0 411.8			28.00		2.20	8.2
010576	SLIME HOLE	3.437	2.8		1340.		3405.	o. o.				200.0 989.9			33.00		2.70	13.4
011276	SLIME HOLE	2.865	2.9		1280.		2355.	0.				146.0 602.2			30.00		2.10	8.7
011976	SLIME HOLE	1.855	2.9		1180.		1403.	0.				155.0 414.1		2124.		77.5	2.10	5.6
012676	SLIME HOLE	1.376	2.9	_	1040.		912.	0.				118.0 233.8		1338.		55.5	2.00	4.0
020276	SLIME HOLE	1.500	2.8		1100.	416.	899.	0.				128.0 276.5		1400.		62.6	2.00	4.3
020976	-1 -4- 11-1-	4 707	2.8		1020.		863.	0.				115.0 230.7		1294.			1.90	3.8
060710	STIME HOLE	1.343	C • O	3.1	10204	430.	003.	U.	U.	130.0	£00.0	113.0 630.7	047.	1674.	67.00	JU . Z	1675	J . U

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LOCATION 0011 SLIME HOLE

DATE	IDENT	FLOW CMM	FIELD LA	SPEC B COND	ACIDITY MG/L KG/D	ALKALINITY MG/L KG/			SULFATE MG/L KG/D	ALUMINUM MG/L KG/D	MANGANESE MG/L KG/D
021676 022376 030176 030876 031576 032276 032276 040576 041276 041276 042676 050376	SLIME HOLE SLIME HOLE SLIME HOLE SLIME HOLE SLIME HOLE	2.046 1.563 1.206 1.470 1.640 1.714 1.458 1.920 1.570 1.293 1.288	3.0 2. 2.8 2. 2.8 2. 2.7 2. 3.0 2. 3.0 2. 2.7 2. 2.7 2. 2.9 2. 2.7 2.	0 1380. 8 1340. 8 1160. 7 1100. 9 1020. 8 950. 7 1080. 7 1100. 9 940. 9 1000. 9 1080.	531. 1564. 550. 1238. 425. 738. 410. 868. 373. 881. 371. 916. 378. 793. 405. 1120. 390. 882. 376. 697. 368. 826.	0. 0 0. 0 0. 0 0. 0 0. 0	. 160.0 471.3 . 170.0 382.6 . 145.0 251.9 . 120.0 254.0 . 110.0 271.5 . 105.0 220.4 . 115.0 317.9 . 100.0 226.1 . 100.0 186.2 . 98.0 181.7 . 100.0 224.4	140.0 315.1 115.0 199.8 98.0 207.4 107.0 252.6 105.0 259.2 105.0 220.4 91.0 251.6 95.0 214.8 98.0 182.5 98.0 181.7	825. 1857. 750. 1303. 710. 1503. 675. 1594. 670. 1654. 675. 1417. 680. 1880. 620. 1402. 605. 1126. 580. 1076.	31.00 69.8 26.00 45.2 27.00 53.7 27.00 66.7 27.00 56.7 29.00 80.2 28.00 63.3 29.00 54.0 28.00 51.9	2.20 6.5 2.10 4.7 1.90 3.3 1.80 3.8 1.80 4.2 1.90 4.7 1.80 3.8 2.00 5.5 1.80 4.1 1.80 3.4 1.80 3.3

AVERAGES FOR 52 SAMPLINGS,

LOCATION DO12 LAUREL RUN

		FLOW	Р	н	SPEC	ACIDITY	ALKAL	INITY	IRON (TOTAL)	IRON (FERR)	SUI	LFATE	ALUM	INUM	MANG	ANESE
		CMM	FIELD	LAB	COND	MG/L KG/	D MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
DATE	IDENT																	
051475	LAUREL RUN	27.507	4.5	4.2	81.	39. 1545	. 0.	0.	0.7	25.7	0.5	17.8	22.	871.	1.00	39.6	0.21	8.3
052175	LAUREL RUN	26.963		4.6	59.	42. 1631		0.	0.6	22.1	0.3	10.5	17.	660.	0.64	24.8	0.20	7.8
052875	LAUREL RUN	13.082		4.8	63.	42. 791	. 0.	0.	0.8	16.0	0.9	16.2	19.	358.	0.84	15.8	0.16	3.0
061175	LAUREL RUN	15.869		4.4	61.	43. 983	. 0.	0.	0.6	13.7	0.4	10.1	19.	434.	0.75	17.1	0.25	5.7
062475	LAUREL RUN	6.066		4.1	79.	40. 349	. 0.	0.	1.0	8.6	0.4	3.3	22.	192.	0.75	6.6	0.34	3.0
070775	LAUREL RUN	3.568	5.0	4.1	94.	65. 334	. 0.	0.	1.6	8.2	1.2	6.2	41.	211.	1.10	5.7	0.42	2.2
072175	LAUREL RUN	4.961	4.7	4.1	97.	62. 443	. 0.	0.	1.2	8.6	1.0	7.1	29.	207.	0.00	0.0	0.00	0.0
080475	LAUREL RUN	2.039	4.3	3.8	133.	58. 170	. 0.	0.	1.2	3.5	1.0	2.9	50.	147.	0.00	0.0	0.00	0.0
081875	LAUREL RUN	73.941	4.7	4.6	58.	40. 4259	. 0.	0.	0.5	54.3	0.4	46.8	16.	1704.	0.00	0.0	0.00	0.0
090275	LAUREL RUN	140.764	4.6	. 4.3	69.	47. 9527	. 0.	0.	0.7	143.9	0.2	32.4	20.	4054.	0.00	0.0	0.00	0.0
091675	LAUREL RUN	14.204	4.5	4.6	69.	36. 736	. 0.	0.	0.7	14.1	0.5	10.2	22.	450.	0.00	0.0	0.00	0.0
092975	LAUREL RUN	23.888	4.1	4.3	86.	45. 1548	. 0.	0.	0.9	31.0	0.6	20.6	24.	826.	0.00	0.0	0.00	0.0
101375	LAUREL RUN	16.565	4.3	4.6	66.	26. 620		0.	0.6	15.0	0.4	10.5	23.	549.	0.00	0.0	0.00	0.0
102775	LAUREL RUN	18.213	4.0	4.7	59.	16. 420	. 0.	0.	0.5	13.1	0.4	10.5	19.	498.	0.00	0.0	0.00	0.0
111075	LAUREL RUN	11.944	4.1	4.6	66.	20. 344		0.	0.8	13.8	0.4	6.9	21.	361.	0.00	0.0	0.00	0.0
112475	LAUREL RUN	18.638	4.0	4.9	55.	9. 242		0.	0.6	16.1	0.4	10.7	17.	456.	0.00	0.0	0.00	0.0
120875	LAUREL RUN	22.716	4.6	4.8	55.	15. 491	. 0.	0.	0.6	19.0	0.4	13.7	17.	556.	0.00	0.0	0.00	0.0
122275	LAUREL RUN	25.230	3.9	4.4	73.	19. 690	. 0.	0.	0.7	25.8		14.5	24.	872.	0.00	0.0	0.00	0.0
010576	LAUREL RUN	103.164	4.1	4.2	113.	48. 7131	. 0.	0.	1.4	208.0	1.0 1	48.6	34.	5051.	0.00	0.0	0.00	0.0
011976	LAUREL RUN		OVER					_										
020276	LAUREL RUN	50.189	3.9	4.3	85.	19. 1373		0.	0.8		0.4			1662.	0.00	0.0	0.00	0.0
021676	LAUREL RUN	208.300	3.5	4.3	77.	15. 4499		0.	0.8	243.0	0.6 1			6599.	0.00	0.0	0.00	0.0
030176	LAUREL RUN	26.845	4.0	4.4	85.	14. 541	. 0.	0.	0.8	31.7	0.4	17.0		1044.	0.00	0.0	0.00	0.0
031576	LAUREL RUN	51.022	4.8	4.8	55.	9. 661		0.	0.4	30.1	0.3	22.0		1249.	0.00	0.0	0.00	0.0
032976	LAUREL RUN	31.466		4.3	86.	15. 680		0.	0.8	34.0	0.4	19.9		1223.	0.00	0.0	0.00	0.0
041276	LAUREL RUN	25.757	4.0	4.5	81.	17. 631		0.	0.8	28.2	0.4	14.1	25.		0.00	0.0	0.00	0.0
042676	LAUREL RUN	46.060	4.2	4.7	78.	16. 1061	. 0.	0.	0.8	53.1	0.4	26.5	23.	1526.	0.00	0.0	0.00	0.0

AVERAGES FOR 27 SAMPLINGS,

FLOW	PH		SPEC	ACIDI	TY	ALKALI	NITY	IRON (T	OTAL)	IRON (FERR)	SUL	FATE	ALUM	INUM	MANG	SANESE
CMM	FIELD	LAB	COND	MG/L I	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
	·		- <i>-</i>			_	•				• •		4257		40.7	0.00	
38.806	4.3	4.4	76.	29. 10	504.	0.	υ.	0.8	43.8	0.5	26.8	22.	1257.	0.55	18.5	0.09	2.0

LOCATION 0013 KILDOW MINE

		FLOW	РН		SPEC	ACID	ITY	ALKALI	NITY	IRON (T	OTAL	IRON (FERR)	\$UL	FATE	ALUM	INUM	MANG	ANESE
		CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L		MG/L	
DATE	IDENT																		
051475	KILDOW MINE	0.005	2.7		576.	110.	1.	o.	Q.	4.2	0.0	0.4	0.0	165.	1.	9.50	0.1	0.18	0.0
052875	KILDOW MINE	0.037	2.5		649.	137.	7.	0.	Q.	5.6	0.3	0.6	0.0	365.	20.	10.00	0.5	0.19	ŏ.ŏ
061175	KILDOW MINE	0.012	3.0		585.	180.	3.	. 0.	0.	8.9	0.2	0.2	0.0	250.	4.	18.00	0.3	0.30	0.0
062475	KILDOW MINE	0.003	2.6		808.	198.	1.	0.	0.	13.0	0.1	0.2	0.0	285.	1.	20.00	0.1	0.40	0.0
070775	KILDOW MINE	0.005	2.8		937.	274.	2.	0.	0.	19.0	0.1	1.2	0.0	330.	2.	28.00	0.2	0.45	0.0
072175	KILDOW MINE	0.007	2.7	2.7 1		377.	4.	0.	0.	28.0	0.3	1.3	0.0	520.	5.	38.00	0.4	0.53	0.0
080475	KILDOW MINE	0.012	2.7	2.6 1		418.	7.	0.	0.	33.0	0.6	2.0	0.0	720.	12.	41.00	0.7	0.62	0.0
081875	KILDOW MINE	0.029	2.8	2.7	840.	235.	10.	0.	0.	14.0	0.6	0.9	0.0	270.	11.	16.00	0.7	0.30	0.0
090275	KILDOW MINE	0.034	2.8	2.8	740.	200.	10.	0.	0.	9.7	0.5	0.8	0.0	223.	11.	14.00	0.7	0.25	0.0
091675	KILDOW MINE	0.008	2.6	2.8	860.	300.	4.	0.	0.	12.0	0.1	1.0	0.0	284.	3.	21.00	0.3	0.31	0.0
092975	KILDOW MINE	0.024	2.8	2.9	752.	192.	7.	0.	0.	8.9	0.3	0.8	0.0	224.	8.	15.00	0.5	0.25	0.0
101375	KILDOW MINE	0.017	2.9	2.8	872.	272.	7.	0.	0.	12.0	0.3	1.4	0.0	283.	7:	21.00	0.5	0.31	0.0
102775	KILDOW MINE	0.012	2.9	2.8	820.	213.	4.	0.	0.	11.0	0.2	1.0	0.0	259.	4.	18.00	0.3	0.25	0.0
111075	KILDOW MINE	0.008	2.8	2.9	825.	248.	3.	0.	0.	12.0	0.1	0.8	0.0	290.		20.00	0.2	0.29	0.0
112475	KILDOW MINE	0.008	2.9	2.8	840.	218.	3.	0.	0.	12.0	0.1	1.0	0.0	266.	3.	19.00	0.2	0.25	0.0
120875	KILDOW MINE	0.008	2.7	2.7	860.	259.	3.	0.	0.	14.0	0.2	0.8	0.0	299.	4.	21.00	0.3	0.30	0.0
122275	KILDOW MINE	0.017	2.5	2.8	712.	176.	4.	0.	0.	9.5	0.2	0.8	0.0	214.		15.00	0.4	0.20	0.0
010576	KILDOW MINE	0.029	2.6		560.	134.	6.	0.	0.	5.5	0.2	0.3	0.0	150.	6.		0.3	0.15	0.0
011976	KILDOW MINE	0.008	3.0		740.	189.	Ž.	ō.	O.	8.8	0.1	0.1	0.0	244.		15.00	0.2	0.21	0.0
020276	KILDOW MINE	0.029	2.6		630.	140.	6.	Ö.	Ō.	4.7	0.2	0.2	0.0	194.		13.00	0.5	0.18	0.0
021676	KILDOW MINE	0.063	2.5		520.	111.	10.	ō.	ō.	4.0	0.4	0.3	0.0	148.		9.10	0.8	0.15	0.0
030176	KILDOW MINE	0.008			630.	141.	Ž.	ō.	Ö.	4.3	0.1	0.3	0.0	194.		12.00	0.1	0.18	0.0
031576	KILDOW MINE	0.063	2.8		700.	166.	15.	Ö.	Ö.	5.9	0.5	0.1	0.0	214.		14.00	1.3	0.20	0.0
032976	KILDOW MINE	0.017	3.0		655.	152.	4.	Ö.	Ö.	5.2	0.1	0.2	0.0	198.		12.00	0.3	0.19	0.0
041276	KILDOW MINE	0.017	2.7		620.	155.	4.	ō.	ō.	5.8	0.1	0.2	0.0	188.		13.00	0.3	0.20	0.0
042676	KILDOW MINE	0.003	2.7		760.	207.	1.	ō.	ů.	11.0	0.1	1.0	0.0	272.		20.00	0.1	0.29	0.0

AVERAGES FOR 26 SAMPLINGS,

LOCATION 0014 KILDOW MINE

		FLOW	PI	н	SPEC	ACID	177	ALKALI	NITY	IRONCT	OTAL)	IRON (FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE
			FIELD	LAB	COND	MG/L	KG/D	M6/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
DATE	IDENT																		
051475	KILDOW MINE	0.008	3.7	3.6	170.	19.	0.	0.	0.	0.2	0.0	0.1	0.0	46-	1.	2.10	0.0	0.05	0.0
052875	KILDOW MINE	0.095	3.3	3.9	177.	23.	3.	0.	0.	0.3	0.0	0.1	0.0	52.	7.	2.70	0.4	0.04	0.0
061175	KILDOW MINE	0.017	3.5	3.6	170.	33.	1.	0.	0.	0.4	0.0	0.1	0.0	58.	1.	3.10	0.1	0.09	0.0
062475	KILDOW MINE	0.008	3.3	3.6	198.	31.	0.	0.	0.	0.6	0.0	0.1	0.0	57.	1.	3.00	0.0	0.20	0.0
070775	KILDOW MINE	0.005	3.8	3.6	225.	45.	0.	0.	0.	1.2	0.0	0.1	0.0	.08	1.	4.60	0.0	0.20	0.0
072175	KILDOW MINE	0.002	3.5	3.5	254.	42.	0.	0.	0.	2.1	0.0	0.1	0.0	100.	0.	5.00	0.0	0.30	0.0
080475	KILDOW MINE	0.002	3.8	3.5	235.	38.	0.	0.	0.	3.3	0.0	2-4	0.0	100.	0.	2.90	0.0	0.39	0.0
081875	KILDOW MINE	0.022	3.3	3.7	175.	31.	1.	0.	0.	0.7	0.0	0.3	0.0	52.	2.	2.30	0.1	0.09	0.0
090275	KILDOW MINE	0.044	3.7	3.7	185.	38.	2.	0.	0.	0.5	0.0	0.1	0.0	59.	4.	2.40	0.2	0.10	0.0
091675	KILDOW MINE	0.008	3.7	3.7	218.	50.	1.	0.	0.	1.1	0.0	0.4	0.0	72.	1.	4.30	0.1	0.15	0.0
092975	KILDOW MINE	0.029	3.8	3.6	216.	36.	1.	0.	0.	0.6	0.0	0.3	0.0	64.	3.	3.50	0.1	0.12	0.0
101375	KILDOW MINE	0.008	3.0	3.6	255.	53.	1.	0.	0.	0.9	0.0	0.1	0.0	74.	1.	4.20	0.1	0.13	0.0
102775	KILDOW MINE	0.024	3.3	3.6	230.	48.	2.	0.	0.	0.6	0.0	0.2	0.0	70.	2.	3.80	0.1	0.10	0.0
111075	KILDOW MINE	0.008	3.8	3.7	235.	47.	1.	0.	0.	1.0	0.0	0.2	0.0	77.	1.	4.40	0.1	0.11	0.0
112475	KILDOW MINE	0.017	2.9	3.5	240.	39.	1.	0.	0.	0.6	0.0	0.2	0.0	72.	2.	3.70	0.1	0.10	0.0
120875	KILDOW MINE	0.017	2.9	3.4	260.	58.	1.	0.	0.	0.7	0.0	0.2	0.0	80.	2.	3.90	0.1	0.09	0.0
122275	KILDOW MINE	0.037	3.0	3.6	211.	40.	2.	0.	0.	0.4	0.0	0.1	0.0	66.	4.	3.40	0.2	0.06	0.0
010576	KILDOW MINE	0.085	2.9	3.7	152.	26.	3.	0.	0.	0.3	0.0	0.1	0.0	35.	4.	1.70	0.2	0.05	0.0
011976	KILDOW MINE	FROZ	EN																
020276	KILDOW MINE	0.044	2.9	3.7	175.	30.	2.	0.	0.	0.4	0.0	0.1	0.0	51.	3.	2.70	0.2	80.0	0.0
021676	KILDOW MINE	0.114	3.2	3.7	165.	27.	4.	0.	0.	0.3	0.1	0.1	0.0	42.	7.	2.00	0.3	0.07	0.0
030176	KILDOW MINE	0.037	3.3	3.7	185.	31.	2.	0.	0.	0.4	0.0	0.1	0.0	54.	3.	2.20	0.1	0.09	0.0
031576	KILDOW MINE		3.2	3.7	189.	32.	1.	0.	0.	0.3	0.0	0.1	0.0	55.	2.	2.30	0.1	0.10	0.0
032976	KILDOW MINE		3.2	3.7	180.	27.	2.	Ö.	0.	0.2	0.0	0.1	0.0	52.	3.	2.20	0.1	0.09	0.0
041276	KILDOW MINE		3.2	3.7	177.	29.	2.	Ŏ.	ō.	0.2	0.0	0.1	0.0	46.	3.	2.40	0.2	0.10	0.0
042676	KILDOW MINE		3.4	3.6	200.	37.	2.	ō.	ō.	0.4	0.0	0.1	0.0	58.	3.	3.40	0.2	0.10	0.0

AVERAGES FOR 26 SAMPLINGS.

FLOW	PI	1	SPEC	ACIDI	ITY	ALKALI	NITY	IRON(T	OTAL)	IRON (FERR)	SUL	FATE	ALUP	INUM	MANG	ANESE
CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
0.030	3.3	3.6	203.	32.	1.	0.	0.	0.4	0.0	0.1	0.0	53.	2.	2.74	0.1	0.09	0.0

LOCATION 0015 LAUREL RUN

		FLOW	Pi	H	SPEC	ACI	DITY	ALKALI	NITY	IRON (T	OTAL)	IRON (FERR)	รบเ	FATE	ALUM	INUM	MANG	ANESE
		CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L		MG/L		MG/L	KG/D
DATE	IDENT																	no/ E	NO/ D
051475	LAUREL RUN	11.044	5.2	4.6	35.	12.	191.	0.	0.	0.2	3.3	0.1	1.6	7.	118.	0.20	3.2	0.00	
052875	LAUREL RUN	6.184	4.8	5.3	26.	39.	347.	0.	0.	0.5	4.0	0.2	2.1	10.	89.	0.30	2.7	0.08 0.14	1.3
061175	LAUREL RUN	8.818	4.7	4.7	25.	11.	140.	0.	0.	0.3	3.2	0.1	0.6	7.	83.	0.15	1.9	0.18	1.2 2.3
062475	LAUREL RUN	2.061	5.4	4.9	28,	16.	47.	0.	0.	0.4	1.2	0.1	0.1	7.	20.	0.38	1.1	0.15	0.4
070775	LAUREL RUN	2.447	5.3	5.5	24.	22.	78.	0.	0.	0.5	1.7	0.1	0.2	8.	27.	0.15	0.5	0.16	0.6
072175	LAUREL RUN	2.311	5.3	4.8	29.	32.	106.	0.	0.	0.6	2.1	0.1	0.2	9.	30.	0.00	0.0	0.00	0.0
080475	LAUREL RUN	0.289	5.4	5.4	26.	25.	10.	0.	0.	0.3	0.1	0.1	0.0	10.	4.	0.00	0.0	0.00	0.0
081875	LAUREL RUN	34.405	5.4	4.7	33.	24.	1189.	0.	0.	0.3	14.4	0.1	5.0	9.	431.	0.00	0.0	0.00	0.0
090275	LAUREL RUN	68.912	4.7	4.6	35.	20.	1985.	0.	0.	0.2	18.9	0.1	5.0	9.	893.	0.00	0.0	0.00	0.0
091675	LAUREL RUN	6.541	4.8	5.0	28.	9.	85.	0.	0.	0.1	1.3	0.1	0.5	7.		0.00	0.0	0.00	0.0
092975	LAUREL RUN	11.214	4.2	4.8	30.	11.	178.	0.	0.	0.1	1.6	0.1	1.6	7.	113.	0.00	0.0	0.00	0.0
101375	LAUREL RUN	6.558	4.0	5.1	29.	5.	47.	0.	0.	0.1	0.8	0.1	0.6	8.	79.	0.00	0.0	0.00	0.0
102775	LAUREL RUN	6.082	4.3	5.3	27.	10.	88.	0.	0.	0.2	1.8	0.1	0.9	8.	70.	0.00	0.0	0.00	0.0
111075	LAUREL RUN	4.485	4.0	5.2	30.	6.	39.	0.	0.	0.3	1.9	0.1	0.6	7.	45.	0.00	0.0	0.00	0.0
112475	LAUREL RUN	6.082	4.5	5.0	30.	9.	79.	Ö.	0.	0.3	2.6	0.1	0.4	7.	64.	0.00	0.0	0.00	0.0
120875	LAUREL RUN	6.966	4.5	5.1	29.	10.	100.	Ö.	o.	0.3	3.0	0.1	0.5	8.	75.	0.00	0.0	0.00	0.0
122275	LAUREL RUN	8.495	3.9	5.3	31.	7.	86.	Ō.	Ō.	0.3	3.9	0.1	0.6	8.	94.	0.00	0.0	0.00	0.0
010576	LAUREL RUN	25.485	3.8	4.8	33.	7.	257.	Ö.	Ö.	0.2	7.7	0.1	2.9	8.	275.	0.00	0.0	0.00	0.0
011976	LAUREL RUN	ICED	OVER					•	•		. • •		- • •	•			0.0	0.00	0.0
020276	LAUREL RUN	8.495	4.2	4.9	33.	7.	86.	0.	0.	0.2	2.2	0.1	0.6	8.	103.	0.00	0.0	0.00	0.0
021676	LAUREL RUN	171.771	3.8	4.7	38.	9.	2226.	Ō.	O.	0.4	86.6	0.1	12.4		2968.	0.00	0.0	0.00	0.0
030176	LAUREL RUN	8.155	4.0	5.4	27.	6.	70.	ō.	Ŏ.	0.3	3.5	0.1	0.9	6.	73.	0.00	0.0	0.00	0.0
031576	LAUREL RUN	8.894	4.3	4.9	30.	9.	115.	0.	O.	0.4	5.0	0.1	0.6	7.	90.	0.00	0.0	0.00	0.0
032976	LAUREL RUN	8.267	4.0	5.5	28.	5.	60.	o.	ō.	0.2	2.4	0.1	0.6	7.	77.	0.00	0.0	0.00	0.0
041276	LAUREL RUN	6.966	4.5	5.2	27.	6.	60.	ō.	Ö.	0.1	1.4	0.1	0.5	6.	63.	0.00	0.0	0.00	0.0
042676	LAUREL RUN	8.240	4.0	5.0	36.	10.	119.	Õ.	0.	0.5	6.3	0.1	1.2	9.	104.	0.00	0.0	0.00	0.0

AVERAGES FOR 26 SAMPLINGS,

FLOW CMM	PI FIELD	•	SPEC	ACID	ITY	ALKALI	NITY	IRON (T	OTAL)	IRON	FERR)	SUL	FATE	ALUM		MANG	ANESE
CMM	LIELD	LAD	COND	MG/L	K670	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
17.567	4.5	5.0	30.	12.	311.	0.	0.	0.3	7.2	0.1	1.6	10-	242	0.07	1 0	0.05	1 2

LOCATION 0016 ARNOLD RUN

		FLOW	PH	1	SPEC	ACID	ITY	ALKALI	NITY	IRON (T	OTAL)	IRON (FERR)	SULF	ATE.	ALUM	INUM	MANG	ANESE
		CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
DATE	IDENT																		
051475	ARNOLD RUN	7.968	4.3	4.1	115.	35.	402.	0.	0.	0.5	5.7	0.3	2.9	30.	344.	1.40	16.1	0.46	5.3
U52875	ARNOLD RUN	2.052	4.2	4.2	141.	48.	142.	0.	0.	0.4	1.2	0.2	0.7	46.	136.	1.80	5.3	0.61	1.8
061175	ARNOLD RUN	1.983	4.5	4.1	139.	43.	123.	0.	0.	0.7	1.9	0.3	1.0	54.	154.	1.50	4.3	0.75	2.1
062475	ARNOLD RUN	0.890	4.1	3.9	147.	48.	62.	0.	0.	0.7	0.8	0.5	0.6	54.	69-	1.90	2.4	0.94	1.2
070775	ARNOLD RUN	0.782	4.0	3.9	196.	42.	47.	0.	0.	0.7	0.8	0.3	0.3	72.	81.	1.80	2.0	1.20	1.4
072175	ARNOLD RUN	1.257	4.7	3.8	175.	48.	87.	0.	0.	0.6	1.1	0.3	0.5	65.	118.	1.70	3.1	1.00	1.8
080475	ARNOLD RUN	0.048	4.3	3.9	175.	38.	3.	0.	0.	0.7	0.0	0.4	0.0	84.	6.	1.50	0.1	1.20	0.1
081875	ARNOLD RUN	11.315	4.0	. 4.1	94.	30.	489.	0.	0.	0.5	8.1	0.2	3.9	27.	440.	0.85	13.9	0.35	5.7
090275	ARNOLD RUN	20.932	4.6	4.2	89.	23.	693.	0.	0.	0.4	12.1	0.3	9.6	26.	784.	0.62	18.7	0.30	9.0
091675	ARNOLD RUN	2.124	4.2	4.2	135.	30.	92.	0.	0.	0.6	1.9	0.3	0.9	47.	144.	1.40	4.3	0.71	2.2
092975	ARNOLD RUN	2.667	3.7	4.1	132.	28.	108.	0.	0.	0.5	1.9	0.2	0.8	44.	169.	1.50	5.8	0.65	2.5
101375	ARNOLD RUN	2.447	3.7	4.2	149.	29.	102.	0.	0.	0.5	1.8	0.5	1.8	48. '	169.	0.00	0.0	0.00	0.0
102775	ARNOLD RUN	2.022	4.0	4.4	120.	18.	52.	0.	0.	0.5	1.5	0.2	0.6	44.	128.	1.20	3.5	0.57	1.7
111075	ARNOLD RUN	1.893	4.0	4.2	147.	19.	52.	0.	0.	0.6	1.6	0.2	0.5	55. '	150.	0.00	0.0	0.00	0.0
112475	ARNOLD RUN	2.124	4.0	4.3	120.	23.	70.	0.	0.	0.7	2.1	0.2	0.6	48. 1	147.	0.00	0.0	0.00	0.0
120875	ARNOLD RUN	2.447	3.7	4.3	111.	22.	78.	0.	0.	0.6	2.3	0.1	0.5	44.	155.	0.00	0.0	0.00	0.0
122275	ARNOLD RUN	3.160	3.9	4.2	123.	20.	91.	0.	0.	0.6	2.7	0.1	0.5	46. 7	209.	0.00	0.0	0.00	0.0
010576	ARNOLD RUN	5.488	3.5	4.1	116.	20.	158.	0.	0.	0.5	3.8	0.1	8.0	35. 2	277.	1.30	10.3	0.42	3.3
011976	ARNOLD RUN	2.124	3.7	3.9	140.	20.	61.	0.	0.	0.5	1.5	0.1	0.4	41. 1	125.	1.40	4.3	0.42	1.3
020276	ARNOLD RUN	2.973	3.6	4.2	128.	22.	94.	0.	0.	0.6	2.6	0.1	0.2	45.	193.	1.60	6.9	0.46	2.0
021676	ARNOLD RUN	26.590	3.7	4.2	85.	9.	345.	0.	0.	0.4	15.3	0.1	2.3	27. 10	034.	0.95	36.4	0.30	11.5
030176	ARNOLD RUN	3.160	4.1	4.3	123.	17.	77.	0.	0.	0.5	2.0	0.1	0.2	41. 1	187.	1.20	5.5	0.40	1.8
031576	ARNOLD RUN	3.972	3.7	4.3	104.	15.	86.	0.	0.	0.4	2.2	0.1	0.3	32. 1	183.	1.00	5.7	0.35	2.0
032976	ARNOLD RUN	2.966	3.7	4.2	100.	19.	81.	0.	0.	0.4	1.7	0.1	0.2	35. 1	150.	0.00	0.0	0.00	0.0
041276	ARNOLD RUN	2.973	3.8	4.3	113.	19.	81.	0.	0.	0.4	1.6	0.1	0.2	38. 1	163.	0.00	0.0	0.00	0.0
042676	ARNOLD RUN	3.551	4.0	4.1	122.	17.	87.	0.	0.	0.7	3.3	0.1	0.5	38. 1	194.	1.20	6.1	0.45	2.3

AVERAGES FOR 26 SAMPLINGS,

FLOW	Pł	1	2 DE C	ACID	ITY	ALKALI	NITY	IRON(T	OTAL)	IRON (FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE
CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
			430		4	•	_		- 4			٠,	~~~	4 22		0.45	- 4
4.612	4.0	4.7	728.	22.	145.	0.	U.	0.5	5.7	0.2	7.2	54.	227.	1.22	8.1	0.47	5.1

LOCATION 0017 LAUREL RUN

		FLOW	Р	Н	SPEC	ACIE	YTIC	ALKALI	NITY	IRON (T	CAATO	I RON (FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE
		CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	
DATE	IDENT																		
052875	LAUREL RUN	4.706	4.9	5.5	37.	36.	244.	0.	0.	0.4	2.9	0.1	0.7	11.	75.	0.35	2.4	0.18	1.2
061175	LAUREL RUN	2.718	4.8	5.0	37.	44.	172.	0.	0.	0.6	2.2	0.4	1.4	11.	43.	0.10	0.4	0.20	0.8
062475	LAUREL RUN	1.145	5.1	5.1	37.	41.	68.	0.	0.	0.9	1.5	0.3	0.5	12.	20.	0.50	0.8	0.25	0.4
070775	LAUREL RUN	0.284	4.9	5.1	42.	50.	20.	0.	0.	1.9	0.8	0.6	0.2	13.	5.	0.45	0.2	0.33	0.1
072175	LAUREL RUN	0.133	5.3	5.2	41.	62.	12.	0.	0.	4.5	0.9	1.2	0.2	19.	4.	0.00	0.0	0.00	0.0
080475	LAUREL RUN	0.056	4.7	4.4	56.	49.	4.	0.	0.	1.2	0.1	0.4	0.0	21.	2.	0.00	0.0	0.00	0.0
081875	LAUREL RUN	15.393	4.7	4.7	42.	32.	709.	0.	0.	0.5	10.4	0.4	8.4	11.	244.	0.00	0.0	0.00	0.0
090275	LAUREL RUN	37.446	4.5	4.9	38.	29.	1564.	0.	0.	0.3	15.6	0.2	8.6	9.	485.	0.00	0.0	0.00	0.0
091675	LAUREL RUN	4.927	5.2	5.1	39.	21.	149.	0.	0.	0.7	4.6	0.3	1.8	12.	85.	0.00	0.0	0.00	0.0
092975	LAUREL RUN	3.738	4.3	5.2	38.	11.	59.	0.	0.	0.4	2.2	0.2	1.1	11.	59.	0.00	0.0	0.00	0.0
101375	LAUREL RUN	6.320	4.3	5.1	40.	10.	91.	0.	0.	0.3	2.6	0.2	1.8	13.	118.	0.00	0.0	0.00	0.0
102775	LAUREL RUN	5.437	4.4	5.2	37.	8.	63.	0.	0.	0.3	2.3	0.1	0.8	11.	86.	0.00	0.0	0.00	0.0
111075	LAUREL RUN	3.364	4.3	4.8	40.	17.	82.	0.	0.	0.5	2.4	0.4	1.9	12.	58.	0.00	0.0	0.00	0.0
112475	LAUREL RUN	1.704	5.0	5.2	38.	9.	22.	0.	0.	0.5	1.2	0.4	1.0	11.	27.	0.00	0.0	0.00	0.0
120875	LAUREL RUN	1.529	4.5	5.2	39.	11.	24.	0.	0.	0.4	0.9	0.2	0.4	11.	24.	0.00	0.0	0.00	0.0
122275	LAUREL RUN	1.893	4.2	5.2	40.	9.	25.	0.	0.	0.3	0.8	0.1	0.3	13.	35.	0.00	0.0	0.00	0.0
010576	LAUREL RUN	4.536	4.3	4.9	41.	9.	59.	0.	0.	0.2	1.6	0.1	0.3	12.	78.	0.00	0.0	0.00	0.0
021676	LAUREL RUN	50.478	4.0	4.9	41.	5.	363.	0.	0.	0.3	21.1	0.1	3.6	11.	800.	0.00	0.0	0.00	0.0
030176	LAUREL RUN	4.757	4.8	5.2	40.	7,	48.	0.	0.	0.3	2.1	0.3	2.1	11.	75.	0.00	0.0	0.00	0.0
031576	LAUREL RUN	12.287	4.4	5.4	37.	7.	124.	0.	0.	0.3	4.4	0.1	0.9	10.	177.	0.00	0.0	0.00	0.0
032976	LAUREL RUN	5.614	4.0	5.2	43.	7.	57.	0.	0.	0.3	2.0	0.1	0.6	11.	89.	0.00	0.0	0.00	0.0
041276	LAUREL RUN	2.277	4.1	5.2	39.	9.	30.	0.	0.	0.3	0.8	0.1	0.3	10.	33.	0.00	0.0	0.00	0.0
042676	LAUREL RUN	3.789	4.4	5.4	38.	14.	76.	0.	0.	0.7	3.5	0.2	1.0	12.	65.	0.00	0.0	0.00	0.0

AVERAGES FOR 23 SAMPLINGS,

LOCATION DODA SNOWY CREEK

		FLOW	P	н	SPEC	ACID	ITY	ALKALI	YTIN	IRONCT	OTAL)	IRON (FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE
		CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
DATE	IDENT																		
							_	_											
051475	SNOWY CREE			6.6	103.	0.	0.	3.	11.	0.7	2.5	0.3	1.1	12.	44.	0.10	0.4	0.16	0.6
052875	SNOWY CREE			6.5	96.	23.	26.	0.	0.	0.6	0.7	0.3	0.3	14.	16.	0.10	0.1	0.20	0.2
061175	SNOWY CREE	0.595	5.5	6.2	123.	17.	15.	0.	0.	0.6	0.5	0.2	0.2	13.	11.	0.10	0.1	0.21	0.2
062475	SNOWY CREEK	0.056	6.2	6.4	162.	7.	1.	0.	0.	1.2	0.1	0.2	0.0	15.	1.	0.10	0.0	0.31	0.0
070775	SNOWY CREE	0.131	6 _ 4	6.6	168.	3.	1.	0.	0.	0.8	0.1	0.2	0.0	16.	3.	0.10	0.0	0.35	0.1
072175	SNOWY CREEK	0.095	6.3	6.7	174.	6.	1.	0.	0.	1.1	0.2	0.3	0.0	14.	2.	0.00	0.0	0.00	0.0
080475	SNOWY CREE!	0.037	6.0	6.7	200.	0.	0.	0.	0.	0.7	0.0	0.2	0.0	13.	1.	0.00	0.0	0.00	0.0
081875	SNOWY CREEK	2.481	5.5	6.2	101.	6.	21.	ο.	0.	0.6	2.1	0.2	0.8	14.	50.	0.00	0.0	0.00	0.0
090275	SNOWY CREEK	0.946	5.3	6.2	89.	5.	7.	0.	0.	0.4	0.5	0.3	0.4	15.	20.	0.00	0.0	0.00	0.0
091675	SNOWY CREEK	0.284	5.8	6.8	141.	6.	2.	0.	0.	0.8	0.3	0.3	0.1	14.	6.	0.00	0.0	0.00	0.0
092975	SNOWY CREEK	0.189	5.4	6.6	115.	3.	1.	0.	0.	0.8	0.2	0.4	0.1	13.	4.	0.00	0.0	0.00	0.0
101375	SNOWY CREEK	0.189	5.7	6.6	140.	4.	1.	0.	0.	0.8	0.2	0.2	0.0	16.	4.	0.00	0.0	0.00	0.0
102775	SNOWY CREEK	0.474	5.8	6.6	113.	3.	2.	0.	0.	0.8	0.5	0.3	0.2	12.	8.	0.00	0.0	0.00	0.0
111075	SNOWY CREEK	0.379	5.5	6.6	156.	0.	0.	0.	0.	1.0	0.5	0.3	0.2	15.	8.	0.00	0.0	0.00	0.0
112475	SNOWY CREEK	0.474	5.5	6.4	115.	2.	1.	0.	0.	0.9	0.6	0.3	0.2	13.	9.	0.00	0.0	0.00	0.0
120875	SNOWY CREEK	0.284	5.7	6.6	114.	0.	0.	3.	1.	1.1	0.4	0.2	0.1	16.	7.	0.00	0.0	0.00	0.0
122275	SNOWY CREEK	0.379	5.6	7.0	99.	0.	0.	7.	4.	0.8	0.4	0.2	0.1	14.	8.	0.00	0.0	0.00	0.0
010576	SNOWY CREEK	0.474	5.6	6.4	100.	0.	0.	0.	0.	0.5	0.3	0.1	0.1	16.	11.	0.00	0.0	0.00	0.0
011976	SNOWY CREEK																		
020276	SNOWY CREEK																		
030176	SNOWY CREEK		5.6	6.2	112.	0.	0.	4.	3.	0.6	0.4	0.2	0.2	14.	10.	0.00	0.0	0.00	0.0
031576	SNOWY CREEK		5.6	6.2	97.	Ö.	Ď.	3.	2.	0.5	0.3	0.2	0.1	13.	7.	0.00	0.0	0.00	0.0
032976	SNOWY CREEK		5.8	6.3	107.	o.	ő.	7.	4.	0.6	0.3	0.2	0.1	13.	7.	0.00	0.0	0.00	0.0
041276	SNOWY CREEK		5.7	6.1	106.	ŏ.	ŏ.	ž.	1.	0.5	0.3	0.2	0.1	13.	9	0.00	0.0	0.00	0.0
042676	SNOWY CREEK		6.0	6.3	130.	Ö.	Ö.	2.	i.	1.0	0.4	0.2	0.1	15.	6.	0.00	0.0	0.00	0.0
5460.0		. 0.207	0.0	0.5		٠.		~ •	• •		5.7			•		2.00		J	

AVERAGES FOR 25 SAMPLINGS,

FLOW	PH	1	SPEC	ACID	ITY	ALKALI	NITY	IRON (T	OTAL)	IRON (FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE
CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
A 643		, ,	437		,	•	•	0.7	0.6	0.2	0.3	4.4	40	0.44	0.4	n 20	0.2

LOCATION OUDC SNOWY CREEK

		FLOW	PI	н	SPEC	AC1D	177	ALKALI	NITY	IRON (T	OTAL	IRON (FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE
		CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D		KG/D
DATE	IDENT																		
051475	SNOWY CREEK	1.103	5.0	6.4	50.	18.	29.	0.	0.	0.5	0.8	0.2	0.4	10.	16.	0.10	0.2	0.09	0.1
052875	SNOWY CREEK	3.058	5.2	6.7	48.	27.	119.	0.	0.	0.6	2.6	0.3	1.1	9.	37.	0.15	0.7	0.19	0.8
061175	SNOWY CREEK	0.644	5.7	6.4	47.	34.	32.	0.	0.	0.6	0.6	0.4	0.3	8.	7.	0.10	0.1	0.19	0.2
062475	SNOWY CREEK	0.683	6.0	6.5	51.	29.	29.	0.	0.	1.0	1.0	0.6	0.6	11.	11.	0.21	0.2	0.20	0.2
070775	SNOWY CREEK	0.340	5.6	6.6	56.	24.	12.	0.	0.	1.4	0.7	0.7	0.3	14.	7.	0.10	0.0	0.30	0.1
072175	SNOWY CREEK	0.652	6.2	6.6	70.	30.	28.	0.	0.	2.3	2.2	1.5	1.4	20.	19.	0.00	0.0	0.00	0.0
080475	SNOWY CREEK	0.056	5.8	6.7	52.	26.	2.	0.	0.	1.4	0.1	0.5	0.0	5.	0.	0.00	0.0	0.00	0.0
081875	SNOWY CREEK	2.990	5.3	6.3	50.	22.	95.	0.	0.	0.7	3.1	0.5	2.0	9.	37.	0.00	0.0	0.00	0.0
090275	SNOWY CREEK	0.780	5.2	6.3	45.	12.	13.	0.	0.	0.7	0.8	0.3	0.3	9.	10.	0.00	0.0	0.00	0.0
091675	SNOWY CREEK	0.496	5.8	6.7	62.	6.	4.	0.	0.	1.7	1.2	0.8	0.6	10.	7.	0.00	0.0	0.00	0.0
092975	SNOWY CREEK	0.785	5.5	6.6	51.	5.	6.	0.	0.	1.3	1.5	0.7	0.8	7.	8.	0.00	0.0	0.00	0.0
101375	SNOWY CREEK	0.620	5.4	6.6	48.	6.	5.	0.	0.	1.0	0.9	0.6	0.5	4.	4.	0.00	0.0	0.00	0.0
102775	SNOWY CREEK	0.415	5.5	6.5	54.	10.	6.	0.	0.	1.6	1.0	0.8	0.5	8.	5.	0.00	0.0	0.00	0.0
111075	SNOWY CREEK	0.486	5.6	6.5	64.	4.	3.	0.	0.	1.9	1.3	0.9	0.6	9.	6.	0.00	0.0	0.00	0.0
112475	SNOWY CREEK	0.486	5.9	6.6	48.	0.	0.	0.	0.	0.9	0.6	0.5	0.4	8.	5.	0.00	0.0	0.00	0.0
120875	SNOWY CREEK	0.510	5.5	6.5	53.	5.	4.	0.	0.	1.1	0.8	0.5	0.4	10.	7.	0.00	0.0	0.00	0.0
122275	SNOWY CREEK	0.476	5.5	6.8	50.	0.	0.	0.	0.	0.7	0.5	0.3	0.2	10.	7.	0.00	0.0	0.00	0.0
010576	SNOWY CREEK	0.758	5.2	6.3	49.	3.	3.	0.	0.	0.7	0.7	0.2	0.2	10.	10.	0.00	0.0	0.00	0.0
011976	SNOWY CREEK	FROZ	EN																
020276	SNOWY CREEK	FROZ	EN																
030176	SNOWY CREEK	0.799	5.2	6.3	48.	1.	1.	0.	0.	0.6	0.7	0.1	0.1	9.	10.	0.10	0.1	0.11	0.1
031576	SNOWY CREEK	2.613	5.4	6.3	50.	3.	11.	o.	0.	0.4	1.3	0.2	0.7	9.	35.	0.00	0.0	0.00	0.0
032976	SNOWY CREEK	1.539	5.7	6.2	58.	7.	16.	0.	0.	1.0	2.2	0.4	1.0	10.	21.	0.00	0.0	0.00	0.0
041276	SNOWY CREEK	1.543	5.6	6.3	50.	1.	2.	Ö.	0.	0.4	0.9	0.1	0.3	8.	18.	0.00	0.0	0.00	0.0
042676	SNOWY CREEK	0.284	5.4	6.4	57.	3.	1.	0.	0.	0.8	0.3	0.3	0.1	11.	4.	0.00	0.0	0.00	0.0

AVERAGES FOR 25 SAMPLINGS.

FLOW	PI	4	SPEC	ACID	174	ALKALI	NITY	IRON (T	OTAL)	IRON (FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE	
CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	
0.885	5.5	6.5	53.	13.	17.	0.	0.	0.8	1.0	0.4	0.5	9.	12.	0.17	0.2	0.21	0.3	

LOCATION OODD LAUREL RUN

		FLOW	Р	н	SPEC	ACID	ITY	ALKALI	NITY	IRON (T	OTAL)	IRON (FERR)	SUL	FATE	ALUF	INUM	MANG	ANESE
DATE	IDENT	CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
DATE	IDENI																		
051475	LAUREL RUN	0.029	5.3	6.4	37.	20.	1.	0.	0.	0.6	0.0	0.1	0.0	11.	0.	0.10	0.0	0.10	0.0
112475	LAUREL RUN	0.284	5.3	6.3	42.	6.	2.	0.	0.	0.3	0.1	0.1	0.0	10.	4.	0.15	0.1	0.11	0.0
120875	LAUREL RUN	0.476	5 -4	6.5	48.	2.	1.	0.	0.	0.5	0.3	0.1	0.0	11.	8.	0.00	0.0	0.00	0.0
122275	LAUREL RUN	0.663	5.3	6.5	40.	3.	3.	0.	0.	0.3	0.3	0.1	0.0	10.	10.	0.00	0.0	0.00	0.0
010576	LAUREL RUN	1.893	4.0	6.2	42.	4.	11.	0.	0.	0.3	0.7	0.1	0.1	10.	27.	0.00	0.0	0.00	0.0
011976	LAUREL RUN	FROZ																	
020276	LAUREL RUN	3.024	5.3	6.1	41.	1.	4.	0.		0.1	0.6	0.1	0.2	10.	44.	0.00	0.0	0.00	0.0
021676	LAUREL RUN	2.272		5.9	39.	3.	10.	0.	0.	0.2	0.5	0.1	0.2	9.	31.	0.00	0.0	0.00	0.0
030176	LAUREL RUN	1.514	5.0	6.1	40.	5.	11.	0.	0.	0.5	1.2	0.1	0.1	10.	21.	0.00	0.0	0.00	0.0
031576	LAUREL RUN	1.325	5.2	6.3	39.	3.	6.	0.	0.	0.3	0.6	0.1	0.1	10.	19.	0.00	0.0	0.00	0.0
032976	LAUREL RUN	0.946		6.2	40.	1.	1.	0.	0.	0.3	0.4	0.1	0.1	10.	13.	0.00	0.0	0.00	0.0
041276	LAUREL RUN	0.567	5.2	6.2	39.	3.	2.	0.	0.	0.3	0.2	0.1	0.0	9.	7.	0.00	0.0	0.00	0.0
042676	LAUREL RUN	1.135	5.4	6.4	51.	4.	7.	0.	0.	0.6	1.0	0.1	0.1	13.	21.	0.00	0.0	0.00	0.0

AVERAGES FOR 13 SAMPLINGS.

FLOW	PI	H	SPEC	ACID	ITY	ALKALI	NITY	IRON(T	OTAL)	IRON	(FERR)	SUL	FATE	ALUP	MUNI	MANG	ANESE
CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
1 087	5 1	6 3	42	₹.	5	n	Λ	ΩZ	0.5	n 1	0.1	10	16	0.02	0.0	0-02	0.0

LOCATION COOR YELLOW BOY BH

		FLOW	FIELD	PH LAB	SPEC COND		DITY	ALKALI MG/L		IRON(T			(FERR) KG/D		LFATE KG/D		IINUM KG/D		GANESE KG/D
DATE	IDENT	• • • • • • • • • • • • • • • • • • • •				1107 C	KG/ D	MG/ C	KG/U	MG/L	NG/D	nu/L	KG/U	HG/L	KG/D	MOIL	KG/V	MQ/L	NOTO
051475	YELLOW BOY BI	H DRY																	
052175	YELLOW BOY BI																		
052875	YELLOW BOY BI																		
060475	YELLOW BOY BI	H DRY																	
061175	YELLOW BOY BI																		
061675	YELLOW BOY BI	1 DRY																	
062475	YELLOW BOY BI	i DRY																	
063075	YELLOW BOY BI	+ DRY																	
070775	YELLOW BOY BI																		
071475	YELLOW BOY BI																		
072175	YELLOW BOY BI																		
072875	YELLOW BOY BI																		
080475	YELLOW BOY BI																		
081175	YELLOW BOY BI																		
081875	TELLOW BOY BI																		
082575	YELLOW BOY BI																		
090275	YELLOW BOY BI																		
090875	YELLOW BOY BI																		
091675	YELLOW BOY BI																		
092275	YELLOW BOY BI																		
092975	YELLOW BOY BI																		
100675	AETTOM BOA BI																		
101375	YELLOW BOY BI																		
102075	YELLOW BOY BE																		
102775	YELLOW BOY BI																		
110375	YELLOW BOY BE																		
111075	YELLOW BOY BI																		
111775	YELLOW BOY BI																		
112475	YELLOW BOY BI																		
120175	YELLOW BOY B																		
120875	YELLOW BOY BE																		
121575	YELLOW BOY BE																		
122275	YELLOW BOY BI																		
010576	YELLOW BOY B																		
020276	YELLOW BOY BI																		
042676	AETTOM BOA BI	I DRY																	

AVERAGES FOR 36 SAMPLINGS.

LOCATION DODE GLORY HOLE

		FLOW	PI		SPEC	ACIO	TTY	ALKALI	NITY	IRON(T	OTAL)	IRON	(FERR)	SUI	LFATE	ALUM	INUM	MANG	ANESE
			FIELD	LAB	COND			MG/L					KG/D		KG/D		KG/D	MG/L	
DATE	IDENT				Como	.,												_	
051475	GLORY HOLE	2.326	2.7	3.1	640.	110.	368.	0.		12.0			15.7			13.00		1.90	6.4
652175	GLORY HOLE	0.095	2.7	3.1	624.	114.	16.	0.	0.	12.0	1.6	6.2	0.8	265.	36.	13.00	1.8	1.60	0.2
652875	GLORY HOLE	DRY																	
060475	GLORY HOLE	DRY																	
061175	GLORY HOLE	DRY																	
061675	GLORY HOLE	DRY																	
062475	GLORY HOLE	DRY																	
063075	GLORY HOLE	DRY																	
670775	GLORY HOLE	DRY																	
071475	GLORY HOLE	DRY																	
072175	GLORY HOLE	DRY																	
072875	GLORY HOLE	DRY																	
080475	GLORY HOLE	DRY																	
081175	GLORY HOLE	DRY																	
081875	GLORY HOLE	DRY																	
082575	GLORY HOLE	1.736	2.7	3.0	700.		345.	ο.	0.	16.0		1.8	4.5		675.		35.0	1.70	4.3
090275	GLORY HOLE	3.568	2.7	3.0	810.	182.	935.	0.	0.	23.0			118.2		1536.		77.1	1.60	8.2
090875	GLORY HOLE	1.495	2.7	3.1	720.	150.	323.	0.	0.	18.0	38.8	18.0	38.8	288.		14.00	30.1	1.60	3.4
091675	GLORY HOLE	0.014	2.9	3.2	572.	133.	3.	ο.	0.	7.6	0.1	5.3	0.1	206.		12.00	0.2	1.70	0.0
092275	GLORY HOLE	0.200	3.0	3.3	590.	164.	47.	0.	0.	15.0	4.3	15.0	4.3	250.		13.00	3.8	1.70	0.5
092975	GLORY HOLE	1.098	2.8	3.0	752.	205.	324.	O.	0.	20.0	31_6	9.7	15.3	294.		17.00	26.9	1.90	3.0
100675	GLORY HOLE	0.126	2.7	3.2	630.	157.	28.	θ.	0.	14.0	2.5	6.8	1.2	244.	44.	14.00	2.5	1.60	0.3
101375	GLORY HOLE	DRY																	
102075	GLORY HOLE	DRY																	
102775	GLORY HOLE	DRY																	
110375	GLORY HOLE	DRY																	
111075	GLORY HOLE	DRY																	
111775	GLORY HOLE	DR Y																	
112475	GLORY HOLE	DRY																	
120175	GLORY HOLE	DRY																	
120875	GLORY HOLE	DRY																	
121575	GLORY HOLE	DRY																	
122275	GLORY HOLE	0.629	2.8	3.1	700.		170.	0.		24.0			15.8	304.		16.00	14.5	2.00	1.8
122975	GLORY HOLE	0.693	2.8	3.2	652.		164.	0.	0.	22.0			22.0		289.		13.0	1.80	1.8
010576	GLORY HOLE	6.254	2.8	3.1	735.		1639.	0.	0.		207.1		180.1			16.00		1.80	16.2
011276	GLORY HOLE	1.400	2.9	3.2	685.		335.	0.	0.	23.0	46.4	18.0	36.3	320.		14.00	28.2	1.50	3.0
012676	GLORY HOLE	0.187	.2.9	3.3	554.	146.	39.	0.	0.	18.0	4.8	12.0	3.2	274.	74.	12.00	3.2	1.30	0.3
020276	GLORY HOLE	DRY																	
021676	GLORY HOLE	3.028	3.0	3.2		154.		0.		19.0		10.0				16.00	69.8	1-40	6.1
022376	GLORY-HOLE	2.272	2.9	3.0	670.	158.	517.	0.	0.	21.0	68.7	10.0	32.7	310.	1014.	14.00	45.8	1.40	4.6

LOCATION DOOF GLORY HOLE

		FLOW PH		SPEC	ACID	ITY	ALKALI	NITY	IRON (T	OTAL)	IRON	FERR)	SUL	FATE	ALUP	INUM	MANG	ANESE	
		CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
DATE	IDENT																		
030176	GLORY HOLE	1.514	2.7	3.1	700.	138.	301.	0.	0.	20.0	43.6	11.0	24.0	290.	632.	12.00	26.2	1.30	2.8
030876	GLORY HOLE	DRY																	
031576	GLORY HOLE	0.758	2.8	3.1	660.	129.	141.	0.	0.	18.0	19.6	11.0	12.0	268.	292.	12.00	13.1	1.30	1.4
032976	GLORY HOLE	0.758	2.7	3.0	645.	133.	145.	0.	0.	18.0	19.6	12.0	13.1	272.	297.	12.00	13.1	1.30	1 - 4
040576	GLORY HOLE	1.893	2.7	3.0	620.	130.	354.	0.	0.	17.0	46.3	6.7	18.3	258.	703.	13.00	35.4	1.30	3.5
041276	GLORY HOLE	0.758	3.2	3.1	585.	134.	146.	0.	0.	15.0	16.4	6.0	6.5	244.	266.	13.00	14.2	1.20	1.3
041976	GLORY HOLE	0.019	3.4	3.2	470.	98.	3.	0.	0.	3.3	0.1	2.4	0.1	172.	5.	11.00	0.3	1.60	0.0
042676	GLORY HOLE	DRY																	
050376	GLORY HOLE	0.019	3.0	3.4	440.	89.	2.	0.	0.	1.9	0.1	0.9	0.0	156.	4.	10.00	0.3	1.60	0.0

AVERAGES FOR 49 SAMPLINGS,

LOCATION GOOM ARNOLD MINE

		FLOW	P	н -	SPEC	ACID	ITY	ALKALI	NITY	IRON(T	OTAL)	IRON (FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE
			FIELD	LAB	COND		KG/D		KG/D		KG/D	MG/L	KG/D	MG/L		MG/L	-	MG/L	KG/D
DATE	IDENT																	. •	
051475	ARNOLD MINE	0.019	3.0	3.0	596.	91.	2.	0.	0.	1.3	0.0	0.1	0.0	155.	4.	6.60	0.2	0.46	0.0
052875	ARNOLD MINE	0.114	2.7	3.1	676.	120.	20.	0.	0.	1.6	0.3	0.1	0.0	180.		8.50	1.4	0.69	0.1
061175	ARNOLD MINE	0.029	3.0	3.0	642.	136.	6.	0.	0.	1.9	0.1	0.1	0.0	225.		14.00	0.6	.0.88	0.0
062475	ARNOLD MINE	0.024	2.6	2.9	734.	147.	5.	0.	0.	2.4	0.1	0.1	0.0	245.		15.00	0.5	0.98	0.0
070775	ARNOLD MINE	0.003	2.6	2.8	765.	177.	1.	0.	0.	2.9	0.0	0.1	0.0	254.		20.00	0.1	1.20	0.0
072175	ARNOLD MINE	DRY																	
080475	ARNOLD MINE	DRY																	
081875	ARNOLD MINE	0.076	2.7	2.9	640.	120.	13.	0.	0.	1.4	0.2	0.2	0.0	164.	18.	8.00	0.9	0.81	0.1
082575	ARNOLD MINE	0.051	2.6	2.9	660.	153.	11.	0.	0.	2.0	0.1	0.1	0.0	205.	15.	11.00	0.8	0.88	0.1
090275	ARNOLD MINE	0.183	2.7	2.9	640.	136.	36.	0.	0.	1.5	0.4	0.1	0.0	168.		8.20	2.2	0.69	0.2
091675	ARNOLD MINE	0.041	3.0	2.9	780.	165.	10.	0.	0.	2.5	0.1	0.1	0.0	240.	14.	16.00	0.9	0.93	0.1
092975	ARNOLD MINE	0.075	2.9	2.8	747.	191.	21.	0.	0.	4.1	0.4	0.1	0.0	234.	25.	15.00	1.6	0.88	0.1
101375	ARNOLD MINE	0.017	2.7	2.8	880.	263.	6.	0.	0.	4.8	0.1	0.2	0.0	279.	7.	18.00	0.4	0.96	0.0
102775	ARNOLD MINE	0.012	2.7	2.8	885.	217.	4.	0.	0.	3.7	0.1	0.2	0.0	299.		19.00	0.3	1.00	0.0
111075	ARNOLD MINE	0.003	3.0	2.8	920.	235.	1.	0.	0.	3.9	0.0	0.1	0.0	323.	_	21.00	0.1	1.10	0.0
112475	ARNOLD MINE	DRY						`.							- •				
120875	ARNOLD MINE	0.007	2.5	2.7	920.	249.	2.	0.	0.	4.2	0.0	0.2	0.0	339.	3.	22.00	0.2	1.20	0.0
122275	ARNOLD MINE	0.029	2.5	2.8	760.	164.	7.	0.	0.	2.6	0.1	0.1	0.0	224.	9.	11.00	0.5	0.73	0.0
010576	ARNOLD MINE	0.347	2.8	3.0	580.	148.	74.	0.	0.	2.0	1.0	0.1	0.0	160.	80.	8.80	4.4	0.60	0.3
011976	ARNOLD MINE	FROZE																	
012676	ARNOLD MINE	0.044	2.8	3.0	790.	177.	11.	0.	0.	2.5	0.2	0.1	0.0	244.	16.	13.00	0.8	0.71	0.0
020276	ARNOLD MINE	0.148	2.7	3.0	635.	135.	29.	0.	0.	1.9	0.4	0.1	0.0	194.	41.	11.00	2.3	0.65	0.1
021676	ARNOLD MINE	0.347	2.7	3.1	530.	98.	49.	0.	0.	1.2	0.6	0.1	0.0	144.	72.	8.00	4.0	0.56	0.3
030176	ARNOLD MINE	0.129	8.5	3.0	580.	98.	18.	0.	0.	1.1	0.2	0.1	0.0	172.	32.	6.10	1.1	0.42	0.1
031576	ARNOLD MINE	0.100	2.9	3.0	630.	119.	17.	0.	0.	1.5	0.2	0.1	0.0	176.	25.	10.00	1.4	0.58	0.1
032976	ARNOLD MINE	0.148	2.8	2.9	570.	100.	21.	0.	0.	1.2	0.3	0.1	0.0	168.	36.	7.30	1.6	0.42	0.1
041276	ARNOLD MINE	0.114	2.6	3.1	565.	103.	17.	0.	0.	1.3	0.2	0.1	0.0	156.	26.	7.00	1.1	0.39	0.1
042676	ARNOLD MINE	0.037	2.9	2.8	680.	131.	7.	0.	0.	2.0	0.1	0.1	0.0	204.	11.	12.00	0.6	0.63	0.0

AVERAGES FOR 28 SAMPLINGS.

FLOW	PH	l	SPEC	ACID	ITY	ALKALI	NITY	IRON (T	CLATO	IRON	FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE
CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
0.075	2.8	2.9	700.	129.	14.	0.	0.	1.7	0.2	0.1	0.0	177.	19.	10.89	1.2	0.73	0.1

LOCATION DODO PENDERGAST

		FLOW			SPEC	ACIO		ALKALI		IRON(IRON	FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE
2475	7.05NT	CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D		KG/D
DATE	IDENT																		
052875	PENDERGAST	0.019	2.7		2060.	866.	23.	0.		158.0	4.3	42.0	1.1	1225.	33.	54.00	1_5	1-60	0.0
060475	PENDERGAST	0.008	2.8		2010.	920.	11.	0.		195.0	2.4	30.0	0.4	1450.		59.00	0.7	2.00	ă.č
061175	PENDERGAST	0.019	2.5	2.6	1890.	875.	24.	0.	0.	500.0	5.4	64.0	1.7	1325.	36.	59.00	1.6	2.00	0.1
0,61675	PENDERGAST	0.019	2.5		2010.	911.	25.	0.	0.	220.0	5.9	45.0	1.2	1310.	35.	61.00	1.6	2.00	0.1
062475	PENDERGAST	0.015	2.6		S050°	908.	20.	0.	0.	205.0	4.5	57.0	1.3	1250.	28.	63.00	1 -4	2.20	0.0
063075	PENDERGAST	0.014	2.7	2.5	2040.	877.	17.	0.	0.	185.0	3.6	43.0	0.8	1200.	23.	61.00	1.2	2.00	0.0
070775	PENDERGAST	0.014	2.8	2.5	2020.	890.	17.	0.	0.	190.0	3.7	18.0	0.4	1240.	24.	61.00	1.2	2.10	0.0
071475	PENDERGAST	0.008	2.6		1900.	826.	10.	0.		194.0	2.4	66.0	0.8	1150.	74.	57.00	0.7	1.80	0.0
072175	PENDERGAST	0.008	2.7	2.5	1830.	824.	10.	0.	0.	188.0	2.3	71.0	0.9	1250.	15.	58.00	0.7	1.80	0.0
072875	PENDERGAST	0.007	2.6	2.5	1550.	787.	8.	٥.	0.	180.0	1.8	66.0	0.6	1270.	12.	53.00	0.5	1.70	0.0
080475	PENDERGAST	0.007	2.7	2.6	1750.	842.	8.	0.	0.	187.0	1.8	60.0	0.6	1310.	13.	54.00	0.5	1.90	0.0
081175	PENDERGAST	0.007	2.5	2.4	1980.	948.	9.	0.	0.	185.0	1.8	46.0	0.5	1410.	14.	55.00	0.5	2.00	0.0
081875	PENDERGAST.	0.010	2.5	2.5	1510.	744.	11.	0.	0.	135.0	2.0	61.0	0.9	920.	14.	40.00	0.6	1.40	0.0
082575	PENDERGAST	0.012	2.7	2.6	1340.	661.	11.	0.	0.	96.0	1.6	32.0	0.5	748.	13.	32.00	0.5	1.10	0.0
090275	PENDERGAST	0.008	2.6	2.7	1360.	689.	8.	0.	0.	112.0	7.4	22.0	0.3	760.	9.	33.00	0.4	1.20	0.0
090875	PENDERGAST	0.008	2.5	2.6	1570.	683.	8.	0.	0.	130.0	1.6	33.0	0.4	935.	11.	43.00	0.5	1.50	0.0
091675	PENDERGAST	0.007	2.6	2.7	1660.	793.	8.	0.	0.	142.0	1.4	35.0	0.3	1000.	10.	48.00	0.5	1.70	0.0
092275	PENDERGAST	0.008	2.8	2.7	1430.	710.	9.	0.		105.0	1.3	33.0	0.4	820.	10.	37.00	0.5	1.10	0.0
092975	PENDERGAST	0.008	2.9	2.6	1700.	781.	10.	0.	0.	130.0	1.6	15.0	0.2	1020.	12.	43.00	0.5	1.40	0.0
100675	PENDERGAST	0.008	2.8	2.7	1800.	818.	10.	0.	0.	145.0	1.8	36.0	0.4	1080.	13.	53.00	0.6	1.60	0.0
101375	PENDERGAST	0.012	2.7	2.6	1880.	846.	14.	0.	0.	155.0	2.7	36.0	0.6	1220.	21.	52.00	0.9	1.70	0.0
102075	PENDERGAST	0.012	0.0	2.7	1720.	804.	14.	0.	0.	140.0	2.4	46.0	0.8	1070.	18.	50.00	0.9	1.60	0.0
102775	PENDERGAST	0.008	2.7	2.6	1800.	873.	11.	0.	0.	170.0	2.1	36.0	0.4	1170.	14.	50.00	0.6	1.60	0.0
110375	PENDERGAST	0.008	2.7	2.6	1750.	850.	10.	0.	0.	170.0	2.1	33.0	0.4	1150.	14.	52.00	0.6	1.70	0.0
111075	PENDERGAST	0.008	2.7	2.7	1630.	828.	10.	0.	0.	165.0	2.0	15.0	0.2	1120.	14.	51.00	0.6	1.60	0.0
111775	PENDERGAST	0.008	2.7	2.6	1880.	847.	10.	0.	0.	170.0	2.1	44.0	0.5	1170.	14.	50.00	0.6	1.60	0.0
112475	PENDERGAST	0.008	2.7	2.6	1800.	826.	10.	0.	0.	170.0	2.1	49.0	0.6	1090.	13.	50.00	0.6	1.60	0.0
120175	PENDERGAST	0.008	2.7	2.6	1760.	800.	10.	0.	0.	165.0	2.0	49.0	0.6	1120.	14.	49.00	0.6	1.70	0.0
120875	PENDERGAST	0.012	2.7	2.5	1780.	780.	13.	0.	0.	160.0	2.7	55.0	0.9	1170.	20.	49.00	0.8	1.60	0.0
121575	PENDERGAST	0.012	2.7	2.7	1630.	764.	13.	0.	0.	160.0	2.7	66.0	1.1	1065.	18.	47.00	0.8	1.60	0.0
122275	PENDERGAST	0.017	2.6	2.6	1680.	844.	21.	0.	0.	170.0	4.2	31.0	0.8	1095.		51.00	1.2	1.70	0.0
122975	PENDERGAST	0.012	2.8	2.8	1330.	690.	12.	0.	0.	140.0	2.4	22.0	0.4	970.	17.	46.00	0.8	1.50	0.0
010576	PENDERGAST	0.085	2.7	2.6	1940.	980.	120.	0.	0.	190.0	23.2	15.0	1.8	1250.	153.	59.00	7.2	1.50	0.2
011276	PENDERGAST	0.029	2.9	2.4	2040.	1080.	45.	0.	0.	200.0	8.3	0.6	0.0	1550.	64.	61.00	2.5	1.50	0.1
011976	PENDERGAST	0.017	2.6	2.4	2000.	1110.	27.	0.	0.	230.0	5.6	3.8		1595.		65.00	1.6	1.70	0.0
012676	PENDERGAST	0.017	2.7	2.7	1990.	1170.	29.	0.		240.0	5.9	15.0		1550.		67.00	1.6	1.90	0.0
020276	PENDERGAST	0.044	2.5	2.5	2060.	1180.	75.	0.	0.	245.0	15.6	25.0		1520.		72.00	4.6	1.80	0.1
020976	PENDERGAST	0.029	2.7	2.6	1940.	1060.	44.	0.	0.	235.0	9.8	46.0		1470.		68.00	2.8	1,80	0.1
021676	PENDERGAST	0.100	2.7		1670.	796.	115.	0.		175.0	25.3	2.2		1080.		57.00	8.2	1.20	0.2
022376	PENDERGAST	0.063	2.6		1700.	924.	84.	0.	-	220.0	19.9	2.0		1320.	_	62.00	5.6	1.40	0.1

LOCATION OOOO PENDERGAST

		FLOW	• р	Н	SPEC	ACID	ITY	ALKALI	NITY	IRONCT	OTAL)	IRON	FERR)	SUI	FATE	ALUM	INUM	MANG	ANESE
DATE	IDENT	CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
030176 030876	PENDERGAST PENDERGAST	0.019			1900. 1880.		27. 43.	0. 0.		230.0	6.2 9.2	19.0 55.0		1470. 1520.		658.00	17.7 2.6	1.50	0.0
031576 032276	PENDERGAST PENDERGAST	0.008	2.7	2.6	1710. 1600.	901. 770.	11. 19.	0.		210.0	2.6	68.0 21.0	0.8	1320.	16.	56.00 54.00	0.7	1.50	0.0
032976 040576	PENDERGAST PENDERGAST	0.029	2,7	2.6	1690. 1580.	905. 850.	38. 46.	0. 0.	0.	200.0	8.3 10.0	52.0 40.0		1290.		56.00 55.00	2.3	1.40	0.1
041276 041976	PENDERGAST PENDERGAST	0.017	2.7		1820. 1900.	932. 930.	23. 50.	0.	0.	210.0	5.1	25.0 44.0	0.6	1340.	33.	65.00	1.6	1.50	0.0
042676 050376	PENDERGAST PENDERGAST	0.029 0.029	2.5	2.5		975. 952.	41. 40.	0.		200.0	8.3	38.0 65.0		1330. 1290.	55.	60.00	2.5	1.70 1.60	0.1

AVERAGES FOR 50 SAMPLINGS,

FLOW	PH	l	SPEC	ACID	YTI	ALKALI	NITY	IRON(T	OTAL)	IRON (FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE
CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
0.020	2.6	2.6	1785.	909.	26.	0.	0.	190.6	5.4	30.6	0.9	1252.	35.	69.04	1.9	1.57	0.0

WEST VIRGINIA ACID MINE DRAINAGE STUDY Snowy Creek-Laurel Run Water Sampling Data

LOCATION COOV ARNOLD STRIP

			FLOW	PH	1	SPEC	ACID	ITY	ALKALI	NITY	IRON (T	OTAL)	1RON (FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE
			CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	K6/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
DATE	IDENT																-			
052175	ARNOLD ST	TRIP	0.076	3.4	3.3	304.	40.	4.	0.	0.	0.2	0.0	0.1	0.0	83.	9.	2.90	0.3	0.45	0.0
052875	ARNOLD ST	TRIP	DRY -	•																
061175	ARNOLD ST	TRIP	DRY																	
062475	ARNOLD ST		DRY																	
070775	ARNOLD \$1		DRY																	
072175	ARNOLD ST		DRY																	
080475	ARNOLD \$1		DRY							_					77		3 70			
081875	ARNOLD ST		0.474	2.8	3.1	400.	62.	42.	0.	0.	0.5	0.3	0.1	0.0	77.	53.	2.70	1.8	0.31	0.2
082575	ARNOLD ST		0.284	2.9	3.2	310.	60.	25.	0.	0.	0.3	0.1	0.1	0.0	64. 55.	26.	1.80	0.7	0.28	0.1
090275	ARNOLD ST		0.413	2.7	3.2	300.	49.	29.	0.	0.	0.2	0.1	0.1	0.1		33.	2.10	1.2	0.42	0.2
091675	ARNOLD ST		0.037	3.0	3.2	348.	59.	3.	0.	0.	0.4	0.0	0-1	0.0	72.	4.	3.10	0.2	0.45	0.0
092975	ARNOLD ST		0.129	3.2	2.9	320.	60.	11.	Q.	Q.	0.2	0.0	0.1	0.0	64.	12.	1.90	0.4	0.39	0.1
101375	ARNOLD ST		0.012	2.9	3.1	372.	72.	1.	0.	0.	0.5	0.0	0.1	0.0	82.	1.	2.40	0.0	0.47	0.0
102775	ARNOLD ST		0.037	3.0	3.1	371.	69.	4.	0.	0.	0.5	0.0	0.1	0.0	90.	5.	2.40	0.1	0.40	0.0
111075	ARNOLD ST		DRY																	
112475	ARNOLD ST		DRY					_	_	_					0.3	40	7 70		0.70	
120875	ARNOLD ST		0.075	2.7	2.9	390.	80.	9.	0.	0.	0.5	0.1	0.1	0.0	92.		3.30	0.4	0.38	0.0
122275	ARNOLD ST		0.085	2.7	3.2	324.	49.	6.	0.	0.	0.4	0.0	0.2	0.0	72.	9.	2.20	0.3	0.28	0.0
010576	ARNOLD ST		0.347	2.8	3. 3	285.	34.	17.	0.	0.	0.3	0.1	0.1	0.0	56.	28.	2.00	1.0	0.25	0.1
011976	ARNOLD ST		FROZE					_		_					~~	4.3	2 20		0.40	0.4
012676	ARNOLD ST		0.114	3.0	3.4	326.	50.	8.	0.	0.	0.4	0.1	0.1	0.0	72.		2.20	0.4	0.40	0.1
020276	ARNOLD ST		0.114	2.7	3.3	298.	41.	7.	0.	0.	0.3	0.1	0.1	0.0	66.	11.	2.20	0.4	0.32	0.1 0.3
021676	ARNOLD ST		0.748	3.0	3.4	240.	33.	36.	0.	0.	0.2	0.2	0.1	0.1	52.	56.	1.40	1.5		
030176	ARNOLD ST		0.114	2.9	3.4	255.	35.	6.	0.	0.	0.2	0.0	0.1	0.0	56.	. 9.	1.30	0.2	0.28	0.0
031576	ARNOLD ST		0.347	3.0	3.4	268.	39.	19.	0.	0.	0.3	0.1	0.1	0.0	67.	33.	1.80	0.9	0.30	0.1
032976	ARNOLD ST		0.114	2.8	3.4	254.	37.	6.	0.	0.	0.2	0.0	0.1	0.0	59.	10.	1.70	0.3	0.29	0.0
041276	ARNOLD ST		0.063	2.6	3.4	245.	38.	3.	0.	0.	0.2	0.0	0.1	0.0	56.	5.	1.60	0.1	0.25	0.0
042676	ARNOLD ST	TRIP	0.024	3.0	3.2	340.	49.	2.	0.	0.	0.4	0.0	0.1	0.0	84.	3.	3.40	0.1	0.46	0.0

AVERAGES FOR 28 SAMPLINGS,

FLOW	PI	1	SPEC	ACID	ITY	ALKALI	NITY	IRON (T	OTAL)	I RON (FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE
CMM	FIELD	LAB															
0 430	2 0	, ,	747		0	0		0.7	0.1	0 1	0 0	47	12	2 01	0.5	0.48	0.1

LOCATION GODZ STRIP MINE

		FLOW	PI	н	SPEC	ACID	ITY	ALKALI	NITY	IRON (T	OTAL)	IRON(FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE
DATE	IDENT	CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
DAIL	INEMI											,							
052175	STRIP MINE	0.051	3.2	3.3	521.	78.	6.	0.	0.	0.9	0.1	0.1	0.0	193.	14.	9.00	0.7	1.10	0.1
052875	STRIP MINE	0.056	3.2	3.3	545.	82.	7.	0.	0.	1.1	0.1	0.1	0.0	183.	15.	8.00	0.6	0.95	0.1
061175	STRIP MINE	0.012	3.3	3.2	516.	99.	2.	0.	0.	1.2	0.0	0.2	0.0	205.	4.		0.1	1.20	0.0
062475	STRIP MINE	0.017	2.7	3.1	570.	100.	2.	0.	0.	1.2	0.0	0.1	0.0	230.		10.30	0.3	1.40	0.0
C70775	STRIP MINE	GRAB SM		3.1	590.	112.		o.		1.4		0.1	•••	210.	•	11.00	0.5	1.40	0.0
672175	STRIP MINE	DRY			2.00					• • •		•••				11.00		1.40	
080475	STRIP MINE	DRY																	
081875	STRIP MINE	0.031	2.7	2.9	600.	95.	4.	0.	0.	3.7	0.2	0.4	0.0	175.	R	8.30	0.4	1.20	0.1
082575	STRIP MINE	0.022	2.7	2.9	600.	150.	5.	0.	0.	3.1	0.1	0.4	0.0	207.		10.00	0.3	1.20	0.0
090275	STRIP MINE	0.183	2.7	3.1	560.	107.	28.	Ō.	Ö.	1.9	0.5	0.3	0.1	184.		7.90	2.1	1.00	0.3
091675	STRIP MINE	0.007	3.0	3.2	624.	172.	2.	Ō.	0.	1.9	0.0	0.3	0.0	214.		11.00	0.1	1.30	0.0
092975	STRIP MINE	0.044	3.0	3.2	587.	115.	7.	ō.	Ō.	2.0	0.1	0.4	0.0	209.		10.00	0.6	1.10	0.1
101375	STRIP MINE	0.007	2.8	3.1	646.	124.	1.	ŏ.	õ.	2.1	0.0	0.4	0.0	223.		11.00	0.1	1.30	0.0
102775	STRIP MINE	DRY		J.,	•		• •	•	•		•••	0.4	0.5	223.	۷.	11.00	0.1	1.30	0.0
111075	STRIP MINE	DRY																	
112475	STRIP MINE	DRY																	
120875	STRIP MINE	0.007	2.6	2.9	650.	138.	1.	Ò.	0.	2.5	0.0	0.4	0.0	248.	2.	13.00	0.1	1.20	0.0
122275	STRIP MINE	0.063	2.6	3.1	572.	102.	9.	0.	0.	2.3	0.2	0.3	0.0	219.		8.90	0.8	0.96	0.1
010576	STRIP MINE	0.314	2.8	3.2	470.	82.	37.	ő.	0.	1.3	0.6	0.1	0.0	150.	68.	5.80	2.6	0.85	0.4
011976	STRIP MINE	FROZE	EN .	-		-	-				- • • •		•••		•	J.00		0.03	0.4
012676	STRIP MINE	0.044	3.0	3.3	547.	123.	8.	0.	0.	2.5	0.2	0.2	0.0	194.	12.	8.10	0.5	0.95	0.1
020276	STRIP MINE	0.085	2.6	3.1	505.	92.	11.	0.	Ö.	1.6	0.2	0.1	0.0	174.	21.	7.60	0.9	0.90	0.1
021676	STRIP MINE	0.452	2.9	3.2	400.	64.	42.	O.	o.	1.2	0.8	0.1	0.0	133.	87.	5.80	3.8	0.80	0.5
030176	STRIP MINE	0.063	2.8	3.3	502.	74.	7.	Ö.	o.	1.1	0.1	0.1	0.0	170.	15.	5.80	0.5	0.78	0.1
031576	STRIP MINE	0.075	2.7	3.2	525.	87.	9.	o.	Ö.	1.3	0.1	0.2	0.0	201.	22.	7.40	0.8	0.70	0.1
032976	STRIP MINE	0.063	2.9	3.2	490.	74.	ź.	Ö.	0.	1.1	0.1	0.1	0.0	175.	16.	5.90	0.5	0.79	0.1
041276	STRIP MINE	0.075	2.6	3.2	487.	77.	8.	0.	0.	1.1	0.1	0.1	0.0	168.	18.				
042676	STRIP MINE	0.044	2.8	3.0	570.	92	6.	0.	0.	1.3	0.1	0.1	0.0	204.		6.90	0.7	0.83	0.1
		077			0 •	,	0.	٠.			U.,	0.1	0.0	4 U4 .	13.	10.00	0.6	1.10	0.1

AVERAGES FOR 28 SAMPLINGS,

FLOW	PH							IRON (T							INUM		ANESE
CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
0.063	2 8	7 1	5.4.7	9.5		^		4 6	0.4	0.4	0.0	4.7		0 07		4 47	

LOCATION OUXX ARNOLD STRIP

			FLOW	PH	ı	SPEC	ACID	ITY	ALKALI	NITY	IRON(T	OTAL).	IRON (FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE
			CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D		KG/D	MG/L	
DATE	IDENT																			
052875	ARNOLD :	STRIP	DRY																	
061175	ARNOLD !	STRIP	DRY																	
062475	ARNOLD :	STRIP	DRY																	
070775	ARNOLD :	STRIP	DRY																	
072175	ARNOLD :	STRIP	DRY																	
080475	ARNOLD S	STRIP	DRY																	
081875	ARNOLD S	STRIP	DRY																	
082575	ARNOLD S	STRIP	0.189	2.8	3.1	370.	80.	22.	0.	0.	0.4	0.1	0.1	0.0	74.	20.	2.30	0.6	0.30	0.1
090275	ARNOLD S	STRIP	D.165	2.7	3.3	280.	52.	12.	0.	0.	0.2	0.1	0.1	0.0	54.	13.	1.20	0.3	0.21	0.0
091675	ARNOLD S	STRIP	DRY																	
092975	ARNOLD S	STRIP	DRY																	
102775	ARNOLD !	STRIP	DRY																	
111075	ARNOLD S		DRY																	
112475	ARNOLD S		DRY																	
120875	ARNOLD		DRY																	
122275	ARNOLD		DRY																	
010576	ARNOLD		0.204	2.7	3.3	306.	60.	18.	0.	0.	0.3	0.1	0.1	0.0	64.	19.	1.60	0.5	0.26	0.1
011976	ARNOLD		FROZE								-									
012676	ARNOLD		DRY																	
020276	ARNOLD		DRY																	
021676	ARNOLD		0.807	3.0	3.5	180.	25.	29.	0.	0.	0.3	0.3	0.1	0.1	40.	46.	1.20	1.4	0.21	0.2
030176	ARNOLD		DRY								- • •									
031576	ARNOLD		0.063	3.0	3.3	288.	41.	4.	0.	0.	0.3	0.0	0.1	0.0	57.	5.	1.70	0.2	0.29	0.0
032976	ARNOLD		DRY				• •	. •	-•											
041276	ARNOLD		DRY																	
042676	ARNOLD		DRY																	

AVERAGES FOR 26 SAMPLINGS,

LOCATION 0104 GRAFTON COAL

			FLOW	PI	1	SPEC	ACID	ITY	ALKALI	NITY	IRON (1	OTAL)	IRON (FERR)	SUL	FATE	ALUM	INUM	MANG	ANESE
			CMM	FIELD	LAB	COND	MG/L	KG/D	MG/L	KG/D	MG/Ł	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D	MG/L	KG/D
DATE	IDENT																			
092975	GRAFTON	COAL	0.758	4.7	5.0	36.	13.	14.	0.	0.	0.8	0.9	0.3	0.3	9.	10.	0.25	0.3	0.09	0.1
101375	GRAFTON	COAL	0.758	4.0	5.0	37.	9.	10.	0.	0.	0.6	0.7	0.3	0.3	10.	11.	0.15	0.2	0.10	0.1
102775	GRAFTON	COAL	1.135	4.0	4.0	85.	18.	29.	0.	0.	0.7	1.1	0.3	0.5	16.	26.	0.00	0.0	0.00	0.0
111075	GRAFTON	COAL	1.135	5.0	5.0	33.	14.	23.	0.	0.	1.4	2.3	0.1	0.2	9.	15.	0.00	0.0	0.00	0.0
112475	GRAFTON	COAL	0.946	4.2	4.9	35.	17.	23.	0.	0.	0.6	0.8	0.2	0.3	9.	12.	0.20	0.3	0.10	0.1
120875	GRAFTON	COAL	0.951	4.3	4.9	36.	14.	19.	0.	0.	0.6	0.8	0.2	0.2	9.	12.	0.00	0.0	0.00	0.0
122275	GRAFTON	COAL	1.135	3.5	4.4	36.	13.	21.	0.	0.	0.5	0.7	0.2	0.3	10.	16.	0.00	0.0	0.00	0.0
030176	GRAFTON	COAL	0.758	4.2	5.4	40.	5.	5	0.	0.	0.3	0.3	0.1	0.1	10.	11.	0.15	0.2	0.08	0.1
031576	GRAFTON	COAL	1.325	5.3	4.9	39.	7.	13.	0.	0.	0.2	0.4	0.1	0.1	10.	18.	0.15	0.3	0.09	0.2
032976	GRAFTON	COAL	1.135	4.0	4.8	41.	9.	15.	0.	0.	0.2	0.3	0.1	0.1	10.	16.	0.10	0.2	0.09	0.1
041276	GRAFTON	COAL	0.946	4.0	4.7	36.	7.	10.	0.	0.	0.2	0.2	0.1	0.1	9.	13.	0.15	0.2	0.10	0.1
042676	GRAFTON	COAL	1.135	4.0	5.1	36.	10.	16.	0.	0.	1.4	2.3	0.1	0.2	10.	16.	0.00	0.0	0.00	0.0

AVERAGES FOR 12 SAMPLINGS,

041276

WEST VIRGINIA ACID MINE DRAINAGE STUDY SNOWY CREEK-LAUREL RUN WATER SAMPLING DATA

FLOW РΗ SPEC ACIDITY ALKALINITY IRON(TOTAL) IRON(FERR) SULFATE ALUMINUM MANGANESE CMM FIELD LAB COND MG/L KG/D DATE IDENT 092975 77. GRAFTON COAL 0.082 5.9 185. 18. 5.0 2. 0. 0. 0.2 0.30 1.6 1.4 0.2 9. 0.0 3.00 Ō. 101375 GRAFTON COAL 0.066 5.0 0.2 0.2 62. 0.0 Ď.O 6.0 180. 15. 1. 0. 1.6 1.6 6. 0.00 102075 0.3 GRAFTON COAL 0.073 5.4 5.7 204. 16. 2. 0. 0. 1.7 0.2 1.7 0.2 69. 7. 0.25 0.0 2.90 102775 GRAFTON COAL 0.051 5.0 5.2 102. 9. 33. 2. 2.90 0.2 1. 0. 0. 2.1 0.2 1.8 0.1 0.20 0.0 111075 GRAFTON COAL 0.029 5.1 5.8 185. 0. 0. 0. 2.6 0.1 2.6 0.1 74. 3. 0.50 0.0 2.50 0.1 6. 112475 GRAFTON COAL 0.031 5.7 200. 0. 0. 0.1 74. 3. 0.35 0.0 3.20 0.1 5.1 16. 1. 3.1 2.9 0.1 120875 GRAFTON COAL 0.065 5.9 0. 0. 80. 7. 0.30 0.0 3.50 0.3 4.9 194. 5. 0. 2.9 0.3 2.8 0.3 4. 122275 GRAFTON COAL 0.037 5.2 5.9 211. 10. 1. 0. 0. 2.8 0.2 2.1 0.1 79. 0.00 0.0 0.00 0.0 010576 GRAFTON COAL 0.197 0. 0. 0.2 135. 38. 0.00 0.0 0.00 0.0 5.0 5.7 320. 24. 7. 2.0 0.6 0.8 020276 GRAFTON COAL 0.031 4.9 0.1 130. 0.65 5.50 0.2 5.5 295. 20. 0. 0. 0.1 6. 0.0 1. 1.6 1.6 021676 2.2 GRAFTON COAL 0.228 4.7 5.4 360. 15. 5. 0. 0. 1.1 0.4 1.0 0.3 184. 60. 0.90 0.3 6.80 0.7 030176 GRAFTON COAL 0.082 4.5 5.6 306. 19. 2. 0. 0. 1.5 0.2 1.5 0.2 164. 19. 0.75 0.1 5.90 031576 GRAFTON COAL 0.151 4.9 300. 17. 4. 0. 0. 0.3 0.9 0.2 132. 29. 0.55 0.1 5.50 1.2 5.4 1.4 032976 GRAFTON COAL 0.090 330. 19. 0.2 1.5 0.2 155. 20. 0.65 0.1 5.60 0.7 4.7 5.7 2. 0. 0. 1.6

0.

0.

0.

AVERAGES FOR 16 SAMPLINGS.

042676 GRAFTON COAL

GRAFTON COAL

LOCATION 0106 GRAFTON COAL

0.082

0.048

3.8

4.8

5.3

332.

5.2 400.

13.

14.

2.

1.

ALKALINITY IRON(TOTAL) FLOW SPEC ACIDITY IRON(FERR) SULFATE ALUMINUM MANGANESE MG/L KG/D MG/L KG/D MG/L KG/D MG/L KG/D MG/L KG/D CMM FIELD LAB COND MG/L KG/D MG/L KG/D 0. 0.084 5.6 257. 2. 0. 1.8 0.2 1.4 0.2 128. 15. 0.59 0.1 4.92 0.6 4.9 16.

1.6

2.9

0.2

0.2

1.3

2.7

0.2 172.

0.2 184.

20.

13.

0.60

1.50

0.1

0.1

6.40

6.40

0.8

SUMMARY LISTING OF ALL GRAB SAMPLE POINTS

	FLOW PH	SPEC	ACIDITY	ALKALINITY	IRON (TOTAL)	IRON(FERR)	SULFATE	ALUMINUM	MANGANESE
NATE CITE INCHT	CMM FIELD LA	COND	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D
DATE SITE IDENT	CO40 CMOL E 0 4			_				0.40	0.05
051475 000G GRIMMS PO 051475 000H LAUREL RU			14.	0.	0.1	0.1	6.	0.10	0.05
051475 OOOK LAUREL RU			12.	0.	0.1	0.1 0.1	7. 12.	0.10 0.10	0.04 0.02
052175 OOOX ARNOLD ST			22.	0.	0.2	0.1	194.	18.00	2.90
052875 000H LAUREL RU			103.	0.	0.6		77.	0.45	0.20
052875 000K LAUREL RU			25.	0.	0.1	0.1		0.60	0.15
052875 000P ARNOLD RU			41.	0.	0.2	0.1 0.1	16. 7.	0.30	0.13
062475 0018 LAUREL RU			16.	0.	0.2	1.3	13.	0.29	0.03
063075 000B GRAFTON C			45.	0.	3.6 1.0	0.1	84.	0.95	2.50
070775 0004 SNOWY CRE		• -	0.	48.	1.4	0.4	16.	0.95	0.15
070775 0002 STRIP MIN	·		32. 112.	0.	1.4	0.1	210.	11.00	1.40
071475 000G GRIMMS PO			60.	0. 0.	4.6	3.5	16.	0.20	0.43
072175 0004 SNOWY CRE			24.	0.	1.4	0.5	16.	0.00	0.00
072175 0018 LAUREL RU			34.	0.	4.4	1.2	32.	0.00	0.00
072875 00EE	GRAB SMPL 6.6 6.		22.	0.	1.3	0.2	14.	0.00	0.00
072875 00FF	GRAB SMPL 5.7 6.		0.	5.	0.6	0.1	9.	0.00	0.00
072875 OOGG CORE DRIL			12.	0.	0.1	0.1	3.	0.00	0.00
080475 0004 SNOWY CRE			10.	0.	1.2	0.4	17.	0.00	0.00
080475 0018 LAUREL RU			55.	0.	6.2	2.0	23.	0.00	0.00
080475 000H LAUREL RU			30.	0.	0.5	0.1	7.	0.00	0.00
080475 000P ARNOLD RU			48.	ŏ.	0.4	0.1	11.	0.00	0.00
080475 DOCC SNOWY CRE			0.	30.	0.8	0.4	10.	0.00	0.00
081875 0004 SNOWY CRE			31.	0.	1.5	0.7	18.	0.00	0.00
081875 0018 LAUREL RU			36.	ŏ.	0.6	0.3	11.	0.00	0.00
090275 0004 SNOWY CRE	GRAB SMPL 5.2 6.3		21.	ŏ.	1.1	0.4	16.	0.00	0.00
090275 0052	GRAB SMPL 5.0 6.5		3.	0.	0.2	0.1	5.	0.10	0.10
090275 000H LAUREL RU			19.	o.	0.1	0.1	9.	0.00	0.00
090275 000K LAUREL RU	GRAB SMPL 5.3 6.1		16.	Ö.	0.2	0.1	9.	0.00	0.00
090275 DOCC SNOWY CRE			0.	27.	0.4	0.3	9.	0.00	0.00
091675 0004 SNOWY CRE			2.	O.	0.6	0.3	9.	0.00	0.00
091675 0018 LAUREL RU	GRAB SMPL 4.4 5.1		25.	0.	1.4	1.2	11.	0.00	0.00
091675 0054	GRAB SMPL 4.7 5.5	21.	5.	0.	1.4	1.2	40.	0.00	0.00
091675 00A1	GRAB SMPL 6.0 6.4	152.	0.	0.	65.0	28.0	40.	0.00	0.00
091675 000G GRIMS PON	GRAB SMPL 5.0 5.1	26.	26.	0.	0.2	0.1	7.	0.00	0.00
091675 000P ARNOLD RU	GRAB SMPL 3.5 4.5		7.	Ö.	0.1	0.1	9.	0.00	0.00
	GRAB SMPL 4.0 4.7	29.	24.	0.	0.9	0.2	10.	0.15	0.23
	GRAB SMPL 4.0 4.7	52.	9.	0.	0.1	0.1	15.	0.10	0.21
092975 0103 GRAFTON C		39.	12.	0.	0.3	0.2	9.	0.30	0.10
092975 0107 BALT. COA		97.	15.	0.	11.0	6.5	17.	0.00	0.00
100675 0018 LAUREL RU	GRAB SMPL 4.2 4.5	45.	36.	0.	0.8	0.7	12.	0.00	0.00

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SUMMARY LISTING OF ALL GRAB SAMPLE POINTS

	FLOW PH	SPEC			IRON(TOTAL)	IRON(FERR)	SULFATE	ALUMINUM	MANGANESE
	CMM FIELD	LAB COND	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D
DATE SITE IDENT			7/	0.	0.7	0.5	29.	0:00	
100675 000K LAUREL RU		4.1 100.		0.	1.2	0.5	20.	0.00	0.00
101375 0004 SNOWY CRE		6.7 84.	5.	0.	0.2	0.4	9.	0.00	0.00
101375 000H LAUREL RU		5.4 29.	6.	0.	0.1	0.1	"	0.00	0.00
101375 000P ARNOLD RU		4.6 42.		29.	0.4	0.2	10.	0.00	0.00
101375 OOCC SNOWY CRE		7.2 115.	0.	0.	1.0	0.5	4.	0.00	0.00
102775 0018 LAUREL RU	· · · · · · · · · · · · · · · · · · ·	5.0 47.		0.	1.1	0.6	12.	0.00	0.00
102775 0109 POND ABOV		5.0 35.		0.	0.4	0.1	10. 18.	0.00	0.00
102775 000K LAUREL RU		4.8 51.	18.	0.	1.5	0.1	18.	0.00	0.00
111075 0004 SNOWY CRE		6.6 106.	6.	0.	1.4	0.5	-	0.00	0.00
111075 0018 LAUREL RU		5.4 43.	18.		0.2	0.1	12. 7.	0.00	0.00
111075 000G GRIMS PON		5.4 26.	7.	0.				0.00	0.00
111075 000H LAUREL RU		4.9 27.		0.	0.3	0.1	7.	0.00	0.00
111075 000K LAUREL RU		4.4 73.	30.	0.	0.6	0.2	23.	0.00	0.00
111075 000P ARNOLD RU		4.5 39.		0. 75	0.1	0.1	10.	0.00	0.00
111075 DOCC SNOWY CRE		7.2 135.	0.	35.	0.5	0.3	9.	0.00	0.00
111075 YRO1 YOUGH B S	•	6.1 98.	17.	0.	1.6	0.6	31.	0.85	0.10
111075 YROZ YOUGH B S		6.5 69.		10.	0.6	0.2	12.	0.25	0.06
111775 0110 SIMMS MIN	•	2.7 756.	256.	0.	4.4	0.6	204.	15.00	0.41
112475 0018 LAUREL RU	•	5.2 37.	12.	0.	0.6	0.3	10.	0.00	0.00
112475 000P ARNOLD RU		4.6 40.		0.	0.1	0.1	9.	0.00	0.00
112475 DOCC SNOWY CRE	-	7.3 127.	0.	42.	0.2	0.1	9.	0.00	0.00
112475 YRO1 YOUGH B S	•	5.9 78.		0.	1.4	0.4	22.	0.70	0.09
112475 YROZ YOUGH A S		6.8 56-		13.	0.5	0.1	7.	0.15	0.03
120875 0004 SNOWY CRE		6.5 86.		0.	1.4	0.1	20.	0.00	0.00
120875 000G GRIMMS PO		5.0 29.		o.	0.3	0.1	7.	0.00	0.00
120875 YR01 YOUGH B S		6.4 84.	5.	0.	1.3	0.3	19.	0.00	0.00
120875 YROZ YOUGH A S		6.9 70.	0.	8.	0.5	0.1	12.	0.00	0.00
122275 000P ARNOLD RU	GRAB SMPL 3.5	4.4 49.		0.	0.1	0.1	10.	0.00	0.00
122275 YRO1 YOUGH B S	GRAB SMPL 5.0	5.8 86.	11.	0.	1.6	0.3	21.	0.00	0.00
122275 YRO2 YOUGH A S	GRAB SMPL 5.9	6.8 53.	0.	13.	0.4	0.2	8.	0.00	0.00
010576 000K LAUREL RU	GRAB SMPL 4.0	5.8 40.	6.	0.	0.2	0.1	8.	0.00	0.00
010576 000P ARNOLD RU	GRAB SMPL 3.7	5.8 41.	9.	0.	0.1	0.1	10.	0.00	0.00
010576 YR01 YOUGH B S	GRAB SMPL 5.3	5.7 75.	11.	0.	1.7	1.1	20.	0.00	0.00
010576 YRD2 YOUGH A S		6.4 53.	0.	5.	0.7	0.1	9.	0.00	0.00
010576 ODCC SNOWY CRE		6.8 83.	0.	24.	0.4	0.1	10.	0.00	0.00
020276 0004 SNOWY CRE		5.9 114.	7.	0.	1.3	0.1	20.	0.75	0.10
020276 000G GRIMMS PO		5.4 28.		0.	0.2	0.1	7.	0.00	0.00
020276 ODCC SNOWY CRE		6.7 90.	Ö.	26.	0.3	0.1	11.	0.10	0.06
020276 YR01 YOUGH B S		5.4 88.		0.	2.0	1.4	22.	0.90	0.12
020276 YROZ YOUGH A S		6.2 63.		1.	0.5	0.1	9.	0.20	0.07

SUMMARY LISTING OF ALL GRAB SAMPLE POINTS

	FLOW		н	SPEC		ALKALINITY	IRON(TOTAL)	IRON (FERR)	SULFATE	ALUMINUM	MANGANESE
	CMM	FIELD	LAB	COND	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D	MG/L KG/D
DATE SITE IDENT											
021676 0004 SNOWY CRE	GRAB SM	PL 4.8	6.0	68.	5.	0.	1.0	0.1	17.	0.00	0.00
021676 000K LAUREL RU	GRAB SM	PL 4.2	5.6	42.	3.	0.	0.2	0.1	10.	0.00	0.00
021676 000P ARNOLD RU	GRAB SM	PL 3.7	4.5	48.	11.	0.	0.1	0.1	9.	0.00	0.00
021676 YR01 YOUGH B 5	GRAB SM	PL 5.2	5.6	60.	5.	0.	1.0	0.5	19.	0.40	0.05
021676 YROZ YOUGH A S	GRAB SM	PL 6.3	6.4	62.	0.	3.	0.6	0.1	10.	0.25	0.05
030176 0004 SNOWY CRE	GRAB SM	PL 5.2	5.9	82.	10.	o.	1.9	0.2	23.	0.00	0.00
030176 000K LAUREL RU	GRAB SM	PL 4-0	5.2	41.	10.	ō.	0.2	0.1	13.	0.00	0.00
030176 DOOP ARNOLD RU	GRAB SM	PL 4.0	4.8	46.	14.	Ō.	0.1	0.1	10.	0.00	0.00
030176 DOCC SNOWY CRE	GRAB SM	PL 7.1	6.8	118.	0.	20.	0.7	0.3	9.	0.00	0.00
030176 YR01 YOUGH B S	GRAB SM	PL 4.9	5.9	76.	14.	0.	1.8	1.2	22.	0.00	0.00
030176 YROZ YOUGH A S			6.5	57.	0.	1.	0.3	0.1	8.	0.00	0.00
031576 0111 T.D.LAKE	GRAB SM	PL 3.9	4.6	57.	11.	O.	0.3	0.1	15.	0.50	0.19
031576 0112 T.D.LAKE	GRAB SM	PL 4.0	4.6	58.	9.	Ö.	0.3	0.1	15.	0.40	0.20
031576 YRO1 YOUGH B S			6.2	70.	3.	Ö.	1.0	0.4	16.	0.00	0.00
031576 YRD2 YOUGH A S			6.6	58.	o.	3.	0.3	0.1	8.	0.00	0.00
032276 DAAA A COAL SP			3.4	310.	60.	ő.	0.2	0.1	90.	10.00	1.30
032976 000H LAUREL RU			5.0	30.	7.	Ö.	0.2	0.1	8.	0.00	0.00
032976 YR01 YOUGH B S			5.8	77.	8.	Ö.	1.3	0.8	19.	0.00	0.00
032976 YRO2 YOUGH A S	GRAB SM	PL 6.3	6.4	59.	o.	i.	0.3	0.1	9.	0.00	0.00
041276 0004 SNOWY CRE	GRAB SM	PL 5.4	6.1	83.	3.	ò.	1.4	0.3	20.	0.00	0.00
041276 000P ARNOLD RU	GRAB SMI	PL 3.5	4.7	40.	13.	ő.	0.1	0.1	9.	0.00	0.00
041276 DOCC SNOWY CRE	GRAB SMI	PL 7.0	6.6	96.	0.	24.	0.3	0.1	9.	0.00	0.00
041276 000J GOB FIRE			1.9	7100.	5300.			-		20.00	7.60
042676 0004 SNOWY CRE	GRAB SMI	PL 6.2	6.1	98.	1.	ō.	1.3	0.3	19.	0.00	0.00
042676 0018 LAUREL RU	GRAB SMI	L 4.0	4.8	48.	13.	Ö.	0.4	0.2	13.	0.00	0.00
042676 000G GRIMMS PO	GRAB SMI	L 4.1	5.0	27.	7.	Ö.	0.2	0.1	6.	0.00	0.00
042676 000H LAUREL RU			4.9	34.	9.	Ö.	0.3	0.1	8.	0.00	0.00
042676 000K LAUREL RU	GRAB SMI	L 4.0	4.4	66.	13.	ŏ.	0.3	0.1	16.	0.00	0.00
042676 000P ARNOLD RU	GRAB SME	L 4.2	4.8	42.	12.	ŏ.	0.1	0.1	9.	0.00	0.00
042676 DOCC SNOWY CRE	GRAB SMF	L 7.0	6.8	117.	0.	20.	0.5	0.3	11.	0.00	0.00

APPENDIX D

MOVABLE WALL BULKHEAD MINE SEAL*

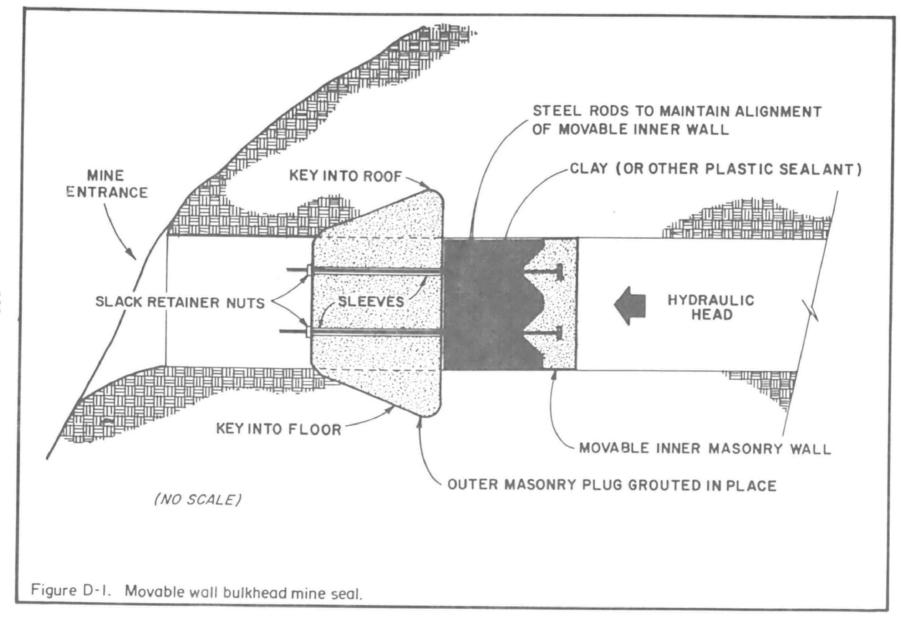
Investigations of double bulkhead mine seals which had failed strongly suggested that some employed seals failed as a result of inability of the rigid keyed inner walls to withstand unanticipated hydraulic pressures. Self-compaction of the central clay core by forces of gravity (including horizontal lattice layering of minerals) over extended intervals of time compounded the problem and often resulted in voids at the roof that render the clay core ineffective as a sealant. Any minor leakage that bypasses the inner wall or occurs upon failure of that wall permits unrestricted water flow directly against the foreward bulkhead and also against the peripheral coal and rock surfaces. The writer is convinced that many failures of traditionally installed double bulkhead seals result from the static force design criteria employed.

The movable wall bulkhead mine seal, illustrated in Figure D-1, is similar in components to the rigid double bulkhead seal having an inner wall, a central core of clay (or other impervious material), and an outer bulkhead (Scott, Robert B. 1972. Evaluation of Bulkhead Seals). Office of Research and Monitoring, U.S. Environmental Protection Agency). However, the proposed mine seal continually utilizes the pressure of the head of water contained to assist in maintaining the integrity of the unit.

In principle, the inner movable wall serves as a ram dynamically driven forward by the changing or constant hydraulic head of water behind the seal. Stress and movement of the wall is transmitted through the central core of clay (or other suitable pliable plastic sealant) pressing that material firmly against all surfaces and weak areas of the center chamber. Bentonite or other swelling clays may be used in the central core area as well as for a thin layer between the inner wall and the mine floor, pillars and roof to insure the integrity and obtain initial compression throughout the core material. The force will continually maintain or improve the efficiency of the clay core for sealing the entry.

In Figure D-1, the inner movable wall is maintained in alignment by means of solid bars that pass through sleeving in the outer fixed or rigid plug. Other alignment arrangements are possible to prevent tilting. The

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illustrated bar connection provides a means of monitoring movement of the inner wall. If the movable wall and bars are designed with sufficient strength, the bars may be used to draw up the wall and effect the initial compression of the core material before development of the hydraulic forces of the water behind the seal.

The outer rigid masonry plug is designed with the strength of a single bulkhead hydraulic seal for the maximum anticipated head of water. This plug should be satisfactorily keyed and grouted into place. Any other standard design may be used. The particular design is similar to one illustrated in 1928 (Zern, E.N., editor, 1928, Coal Miner's Pocketbook, 12th edition, McGraw-Hill, 1928, p.889).

GLOSSARY

- adit: A nearly horizontal entrance to an underground mine.
- anticline: A fold of sedimentary bedrock which dips down and away from a common ridge or axis.
- backfill: Material placed back into an excavation, returning the area to a predetermined contour.
- diversion ditch: Ditch constructed to control surface runoff.
- gob: Mine refuse pile or other coal material removed from the coal through a cleaning process.
- highwall: The deeper exposed face of strata resulting from excavation in surface mining.
- lowwall: The shallower side of excavation in a surface mine cut.
- mine pool: Flooded portions of abandoned deep mine workings.
- mine pool level control lake: A mine pool controlled by the elevation of a surface impoundment.
- outcrop: A natural exposure or position of a geologic unit at the intersection of that unit with the ground.
- overburden: Soil and rock strata overlying a minable mineral.
- pH: Negative logarithm to the base ten of hydrogen ion activity. pH 7 is neutral. Values above pH 7 are basic, those below pH 7.0 is acidic.
- regrade: To change the contour by the use of leveling or grading equipment.
- spoil: Overburden material that is removed as a result of excavating for a marketable mineral.
- syncline: Fold in rocks in which the strata dip inward and downward toward the axis. Opposite of anticline.
- synclinal basin: A basin having characteristics of a syncline.

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15. SUPPLEMENTARY NOTES

This study was conducted in cooperation with West Virginia Department of Natural Resources, Charleston, West Virginia 25311.

16. ABSTRACT

A study was conducted at the Snowy Creek - Laurel Run basin near Terra Alta, West Virginia, to determine the feasibility of demonstrating mine drainage control by known abatement techniques in abandoned coal mine areas having shallow overburden.

The basin contains two abandoned mining complexes that have extensively deepmined the Lower Kittanning coal found in the Mount Carmel syncline. Associated mine pool discharges are responsible for 90 percent of AMD pollution in Snowy Creek which discharges into the Youghiogheny River (now being considered as a part of the National Wild and Scenic Rivers System). Only one-third of the Snowy Creek - Laurel Run basin is affected by AMD.

Additional inundation and stabilization of the mine pools were judged necessary to reduce the AMD pollution. The recommended approach was to utilize continuous clay core dams, a mine pool level control lake and movable wall bulkhead seals to increase the size of the mine pools. It was felt that this abatement approach was feasible.

17. KEY WORDS AND DOCUMENT ANALYSIS											
a. DESCRIPTORS	b.IDENTIFIERS/OPEN ENDED TERMS	c. COSATI Field/Group									
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