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# INDIAN POINT FEASIBILITY STUDY

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## **Prepared for**

**Enercon Services  
Vernon Thompson  
500 TownPark Lane, Suite 275  
Kennesaw, GA 30144-5509**

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## **Prepared by**

**Calvin J. Konya, Ph.D.  
Precision Blasting Services  
P.O. Box 189  
Montville, OH 44064**

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EXECUTIVE SUMMARY

The scope of work was to determine the feasibility and cost of blasting rock for the excavation of the cooling towers for reactors number 2 and 3 and also blasting of the rock trenches for the placement of the new 12 foot diameter pipes for the closed loop cooling system.

The tentative blasting methods were determined and charge weights were calculated. The vibration levels were projected with information in similar rock types. The distances from the excavations to critical structures were determined and vibration levels were calculated at these structures.

Information in the FARC report indicated that the Indian Point 3 facility was built per requirements of the zone 2 of the Uniform Building Code i.e., corresponding to an intensity of VII on the Modified Mercalli Scale.

The maximum safe vibrations levels from blasting or other sources for the entire facility was considered to be no more than an earthquake of an intensity of VII on the Modified Mercalli Scale. The peak particle velocity range which corresponds to a Modified Mercalli Intensity of VII is between 3.5 inches per second to 6.3 inches per second.

A Mercalli Scale intensity of VII would produce a maximum peak particle velocity of about 6.3 inches per second. At a one hertz earthquake frequency this would produce a displacement of 1.114 inches, while a blast vibration producing the same peak particle velocity of 6.3 inches per second at a frequency of 50 hertz would produce a displacement of 0.020 inches. The earthquake ground displacement would be 315 times greater than that of blasting at the same particle velocity.

The review conducted of the past earthquake vibration standards at Indian Point, standards for weakly constructed residential structures, industrial structures, and 80 years of research data proves that blasting can be safely conducted at the reactor without any damage to the structure.

An extremely conservative vibration limit would be 2.0 inches per second which is the limit for residential structures. This vibration limit would protect the structures and allow construction at reasonable costs. According to the available data the reactor should be totally safe at **three times** this vibration limit if in fact it is built to withstand a Mercalli Intensity VII earthquake.

The following measurements were taken from the drawings for the construction of the large diameter pipe installations from the cooling towers. Blasting of trench rock costs \$50/cubic yard. The \$50/yd<sup>3</sup> cost would cover the cost of all types of trench blasting on the project

The size of the trench needed for the four pipes would be 62 feet wide and 20 feet deep and 1722 feet long and equals 79,084 yd<sup>3</sup>. The size of the trench needed for the two pipes would be 40 feet wide and 20 feet deep and 2500 feet long and equals 74,074 yd<sup>3</sup>. The total volume of rock trench equals 153,158 yd<sup>3</sup>. The total cost of trench blasting would be \$7,657,900.

The volume of rock needed for excavation for the towers would be 1,946,842 yd<sup>3</sup> for the two towers. The cost of the excavation for the two cooling towers would be \$7/cubic yard. The cost of the excavation for the towers is 1,946,842 yd<sup>3</sup> times \$7.00 yd<sup>3</sup> equals \$13,627,894.

The total cost of drilling and blasting would include the drilling and blasting cost, blasting consultant cost and seismic monitoring cost. These costs are listed below.

- Drilling and Blasting =  $\$7,657,900 + \$13,627,894 = \$21,285,794$
- Blasting Consultant =  $\$500,000$
- Seismic Monitoring Cost =  $\$600,000$

The total cost of excavation excluding the rock loading and haulage costs would be  $\$22,385,794$ .

There are many different options for the disposal of the  $2,100,000 \text{ yd}^3$  (bank) of rock which will be blasted. The rock can be placed on site, dumped into the ocean, sold to a quarry or crushed and sold from the site.

For the purpose of this study the ocean dumping cost is the highest and the only sure method of disposal since the rock quality is unknown at this time.

The state of New York and the State of New Jersey both are building artificial reefs off their shores. A permit would be needed from the Department of Environmental Conservation (DEC) to ocean dump rock from the Indian Point excavation. This should not be a problem because the DEC encourages the building of these reefs because they attract fish.

The cost of moving the rock from Indian Point to an artificial reef considers a round trip haul distance of 200 miles. Two  $4000 \text{ yd}^3$  barges which hold 4500 tons would be provided as well as an ocean going tug boat. The blasted rock would be loaded into the barge by the construction company at Indian Point. The cost of digging the broken rock and the cost of moving the rock to the barge is not in the price. The cost to dump the  $2,100,000 \text{ yd}^3$  of rock would be  $\$18.90 \text{ per yd}^3$  or  $\$39,690,000$ .

The total cost of drilling, blasting, and disposal (loading and haulage (on land) costs not included) would be  $\$62,075,794$  or  $\$29.56 \text{ yd}^3$ .

## BLASTING FOR ROCK EXCAVATION AT INDIAN POINT

### BACKGROUND

The scope of work was to determine the feasibility of blasting rock for the excavation of the cooling towers for reactors number 2 and 3 and also blasting of the rock trenches for the placement of the new 12 foot diameter pipes for the closed loop cooling system. The initial feasibility study considered the vibration levels generated by the blasting and determined whether the blasting could be done safely. The seismology information concerning the Indian Point site was obtained from the FSAR report.

The tentative blasting methods were determined and charge weights were calculated. The vibration levels were projected with information in similar rock types. The distances from the excavations to critical structures were determined and vibration levels were calculated at these structures.

The cost of excavating rock for placement of the cooling towers and associated trenches were determined as well as cost for disposal of the blasted rock.

### VIBRATION STANDARDS FOR INDIAN POINT

Information in the FARC report indicated that the Indian Point 3 facility was built per requirements of the zone 2 of the Uniform Building Code i.e., corresponding to an intensity of VII on the Modified Mercalli Scale.

The maximum vibrations levels from blasting or other sources therefore for the entire facility will be considered for an earthquake of an intensity of VII on the Modified Mercalli Scale.

A previous study by Consolidated Edison of New York prepared in January 1976 Entitled "Economic and Environment Impact of Alternative Closed Cycle Cooling System" proposed a vibration limit of 1.0 inches per second peak particle velocity for blasting near operating reactors. The report also indicated that blasting was not uncommon at nuclear facilities with active reactors.

### COMPARISON OF MERCALLI AND RICHTER SCALE

The strength of an earthquake is usually measured on one of two scales, the Modified Mercalli Scale and the Richter Scale. The Mercalli Scale is a rather arbitrary set of definitions based upon what people in the area feel, and their observations of damage to buildings around them. The scale goes from 1 to 12, using the descriptive titles of the intensity levels.

The Intensity Scale differs from the **Richter Magnitude Scale** in that the effects of any one earthquake vary greatly from place to place, so there may be many **Intensity** values (e.g.: **IV**, **VII**) measured for the same earthquake. Each earthquake, on the other hand, should have only one **Magnitude**, although the various methods of calculating it may give slightly different values (e.g.: **4.5**, **4.6**). The Richter Scale is designed to allow easier comparison of earthquake magnitudes, regardless of the location.

C.F. Richter was a geologist living and working in California, U.S.A, an area subjected to hundreds of earthquakes every year. He took the existing Mercalli scale and tried to add a 'scientific' scale based on accurate measurements that could be recorded by seismographs (instruments used to measure vibration) regardless of their global location.

By measuring the speed, or acceleration, of the ground when it suddenly moves, he devised a scale that reflects the 'magnitude' of the shock.

The Richter scale for earthquake measurements is logarithmic. This means that each whole number step represents a ten-fold increase in measured amplitude. Thus, a magnitude 7 earthquake is 10 times larger than a 6, 100 times larger than a magnitude 5 and 1000 times as large as a 4 magnitude.

This is an open ended scale since it is based on measurements not descriptions. An earthquake detected only by very sensitive people registers as 3.5 on his scale, while the worst earthquake ever recorded reached 8.9 on the 'Richter Scale'.

Magnitude is a measure of the strength of an earthquake or strain energy released by it, as determined by seismographic observations.

Intensity is a measure of the effects of an earthquake at a particular place on humans, structures and (or) the land itself. The intensity at a point depends not only upon the strength of the earthquake (magnitude) but also upon the distance from the earthquake to the point and the local geology at that point.

There is no direct mathematical relationship between the Mercalli and Richter scale. Scientific observations have been able to make comparisons based on the results and effects. Table 1. shows a comparison of Mercalli Scale and Richter Scale values based on results and effects.

There is no direct general mathematical relationship between Richter scale and peak particle velocity. If some parameters are fixed then peak particle velocity can be calculated for Richter Scale values. The particle velocities are compared for different Richter Scale values in Table 1..

**TABLE 1. COMPARISON OF MERCALLI SCALE AND RICHTER SCALE**

MERCALLI SCALE	RESULTS AND EFFECTS	PPV at 1 Hz	RICHTER SCALE
I	Not noticeable by most humans	0.002 to 0.35	-1 to 3.5
II	Hangings objects sway, noticed by few people	0.35 to 3.5	3.5 to 5.4
III	Many people feel movement	0.35 to 3.5	3.5 to 5.4
IV	Doors windows shelves rattle	0.35 to 3.5	3.5 to 5.4
V	Light fixtures move, pictures fall of walls, objects fall from shelves	0.35 to 3.5	3.5 to 5.4
VI	Light furniture falls over, windows may crack, trees sway	0.35 to 3.5	3.5 to 5.4
VII	Some people fall over, walls may crack	3.5 to 6.3	5.4 to 6.0
VIII	Heavy furniture falls over, some walls may crumble, chimneys fall	7.0 to 17.7	6.1 to 6.9
IX	Some buildings collapse, dams crack,	19.9 to 56	7.0 to 7.9
X	Most buildings damaged, roads crack, railroad tracks bend	Above 62.8	Over 8.0
XI	Bridges collapse, buried pipes break, most buildings collapse	Above 62.8	Over 8.0
XII	All manmade structures are destroyed, total destruction	Above 62.8	Over 8.0

From the three seismic wave parameters; displacement, acceleration, velocity, the velocity is the most closely related to energy and damage potential. This is the reason that the blasting industry uses particle velocity as the indicator for damage to structures.

#### FACTORS AFFECTING BLAST VIBRATION

##### PRINCIPAL FACTORS

There are two principal factors that affect the vibration level that results from detonation of an explosive charge. These are distance and charge size. Common sense indicates that it is safer to be far away from a blast than to be near it. Common sense further indicates that a large explosive charge will be more hazardous than a small charge.



### CHARGE - DISTANCE RELATIONSHIP

Extensive research has been conducted to determine the mathematical relationship between vibration level, charge size, and distance. The U.S. Bureau of Mines Bulletin 656 (Nichols, Johnson and Duvall, 1971) states such a relationship. The relationship is:

$$V = H \left[ \frac{D}{W^\alpha} \right]^\beta$$

where:

- V = Predicted particle velocity (in/s)
- W = Maximum explosive charge weight per delay (lbs)
- D = Distance from shot to sensor measured in 100's of feet (e.g., for distance of 500 feet., D=5)
- H = Particle velocity intercept
- $\alpha$  = Charge weight exponent
- $\beta$  = Slope factor exponent

This is known as the Propagation Law because it shows how the particle velocity changes with distance and explosive charge weight. The numerical values for H  $\alpha$  and  $\beta$  are slightly different for each component. For the longitudinal or radial component, the law is numerically expressed as:

$$V_r = 0.052 \left( \frac{D}{W^{0.512}} \right)^{-1.63}$$

Introducing the following approximations:

$$\begin{aligned} \alpha &= 0.512 \cong 0.5 \\ \beta &= -1.63 \cong -1.6 \end{aligned}$$

Expressing D in feet instead of hundreds of feet produces a simplified approximation for this relationship:

$$V = 100 \left( \frac{d}{\sqrt{W}} \right)^{-1.6}$$

where:

- D = Distance from shot to sensor (ft)  
W = Maximum explosive charge weight per delay (lbs)

The Dupont Blaster's Handbook (E.I. Dupont de Nemours & Co., 1977) gives the following relationship:

$$V = 160 \left( \frac{d}{\sqrt{W}} \right)^{-1.6}$$

#### ESTIMATING PARTICLE VELOCITY

The formulas enable one to estimate the particle velocity likely to result from the detonation of a given charge weight of explosive at a given distance. Obviously the Dupont formula will give a higher value for the expected particle velocity. From this, it can be seen that these formulas serve merely as guides, and only give ballpark figures.

The values of a, b and H are determined by conditions in the area, rock type, local geology, thickness of overburden and other factors. The values of  $\alpha = 0.5$  and  $\beta = 1.6$  are fairly well fixed. The value of H is highly variable and is influenced by many factors.

#### CHARGE WEIGHT, DISTANCE EFFECTS

To illustrate the effect of charge weight and distance, two graphs are presented, one for charge weight vs. particle velocity, the second for distance vs. particle velocity, consider:

$$V = 100 \left[ \frac{d}{\sqrt{W}} \right]^{-1.6} = 100 \frac{W^{0.8}}{d^{1.6}}$$

The above equation is useful in determining vibration levels that would occur at different distances and charge weights from the blast. This equation can be written in a different manner that will help predict the charge weight based on a certain vibration level that we would like to maintain as a maximum value. The equation would be written as follows:

$$W = d^2 \left[ \frac{V}{H} \right]^{1.25}$$

where:

- W = Charge weight per delay (lbs/delay)  
H = 100

$$d = \text{Distance (ft)}$$
$$V = \text{Vibration level}$$

Assuming a charge  $W$  produces a particle velocity  $V$  at a distance,  $d$ , in the equation. Then by letting  $W$  vary in multiples,  $2W$ ,  $3W$ , etc., for the fixed value of  $d$ , the relative values of  $V$  are plotted against charge weight in Figure 1.

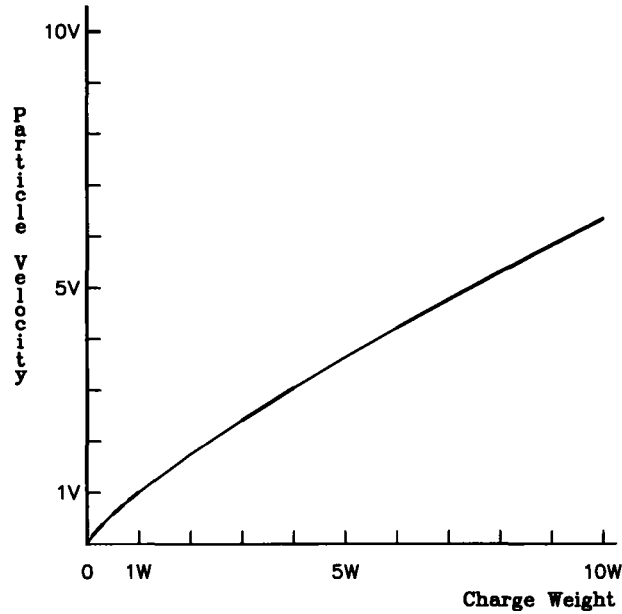
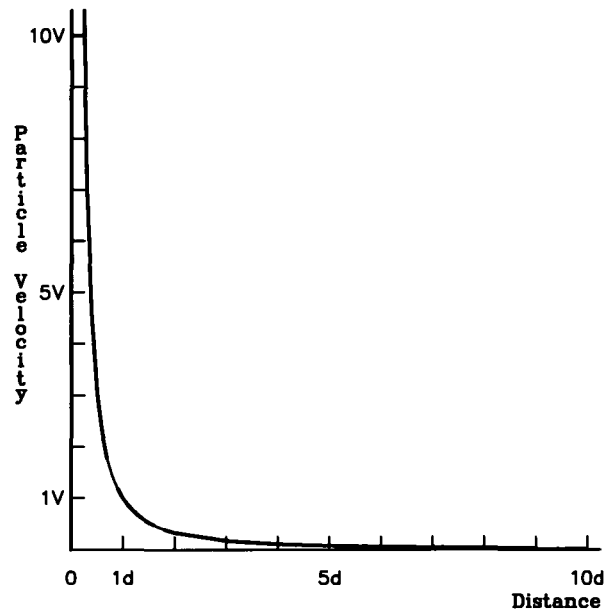


Figure 1. Relative Particle Velocity vs. Charge Weight

Similarly, assuming a charge  $W$  produces a particle velocity  $V$  at a distance  $d$  then by letting the distance increase in multiples  $2d$ ,  $3d$ , etc., for a fixed value of  $W$ , the relative values of  $V$  can be computed. The relative values of  $V$  are plotted against distance in Figure 2.



**Figure 2. Particle Velocity vs. Distance Relationship**

These graphs illustrate the effects produced, but a numerical example will be helpful. Consider the following questions:

Question: If the charge weight is doubled, how much will the particle velocity increase?

Using:  $V_1 = 100 \left( \frac{d}{\sqrt{W}} \right)^{-1.6}$

Calculate:  $V_2 = 100 \left( \frac{d}{\sqrt{2W}} \right)^{-1.6}$

$$V_2 = 100 \left( \frac{d}{1.41\sqrt{W}} \right)^{-1.6}$$

$$V_2 = 1.41^{1.6} 100 \left[ \frac{d}{\sqrt{W}} \right]^{-1.6}$$

$$V_2 = 1.41^{1.6} V_1 = 1.74V_1$$

Answer: Doubling the charge weight increases the particle velocity 1.74 times. (Note: It is not double!)

Question: If the charge weight is cut in half, how much will the particle velocity decrease?

Using:  $V_1 = 100 \left( \frac{d}{\sqrt{W}} \right)^{-1.6}$

Calculate:  $V_2 = 100 \left( \frac{d}{\sqrt{\frac{W}{2}}} \right)^{-1.6}$

$$V_2 = 100 \left( \frac{d}{\frac{1}{1.41} \sqrt{W}} \right)^{-1.6}$$

$$V_2 = \frac{1}{1.41^{-1.6}} 100 \left( \frac{d}{\sqrt{W}} \right)^{-1.6}$$

$$V_2 = \frac{1}{1.74} = 0.57V_1 \cong 0.6V_1$$

Answer: Cutting the charge weight in half will decrease the particle velocity to six tenths its original value. (Note: It is not cut in half!)

Question: If the distance is doubled, how much will the particle velocity decrease?

Using:  $V_1 = 100 \left( \frac{d}{\sqrt{W}} \right)^{-1.6}$

Calculate:  $V_2 = 100 \left( \frac{2d}{\sqrt{W}} \right)^{-1.6}$

$$V_2 = 2^{-1.6} 100 \left( \frac{d}{\sqrt{W}} \right)^{-1.6}$$

$$V_2 = \frac{1}{2^{1.6}} V_1 = 0.33V_1$$

Answer: If the distance is doubled, the particle velocity is reduced to one third of its original value. (Note: It is not cut in half!)

Question: If the distance is cut in half, how much will the particle velocity increase?

Using: 
$$V_1 = 100 \left( \frac{d}{\sqrt{W}} \right)^{-1.6}$$

Calculate: 
$$V_2 = 100 \left( \frac{\frac{d}{2}}{\sqrt{W}} \right)^{-1.6}$$

$$V_2 = 2^{1.6} 100 \left( \frac{d}{\sqrt{W}} \right)^{-1.6}$$

$$V_2 = 2^{1.6} V_1 = 3.03 V_1 \cong 3 V_1$$

Answer: If the distance is cut in half, the particle velocity will be tripled. (Note: It is not doubled!)

#### VIBRATION CONTROL

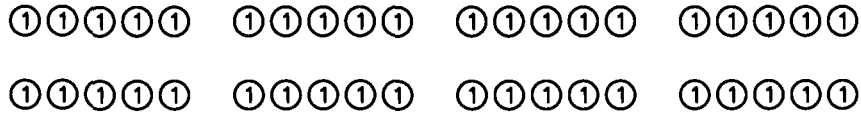
The operator would like to have a convenient, effective means of vibration control. The formulas just discussed are a means to such control, and have led to the development of other techniques.

#### DELAY BLASTING

Before discussing these techniques, delay blasting should be considered. With the development of the delay cap, particularly millisecond delays, a method came into play by which a large explosive charge could be detonated as a series of small charges, rather than one large charge. Obviously, the reduction in charge size can be made by the use of multiple delays. For example, the use of ten delays would reduce the effective vibration generating charge to one tenth the original charge.

**Consider the following example:**

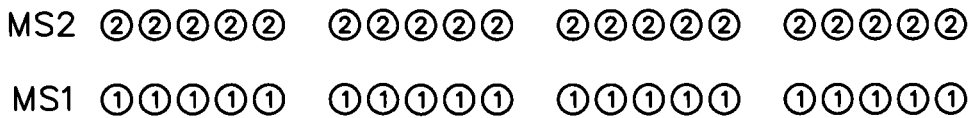
A shot consists of 40 holes, 250 lbs. of explosive per hole with a total charge of 10,000 lbs. and is fired instantaneously. The probable vibration level can be calculated at a distance of 1,000 feet.



40 Holes Fired Instantaneously

$$V = 100 \left( \frac{1000}{\sqrt{10000}} \right)^{-1.6} = 2.51 \text{ in/s}$$

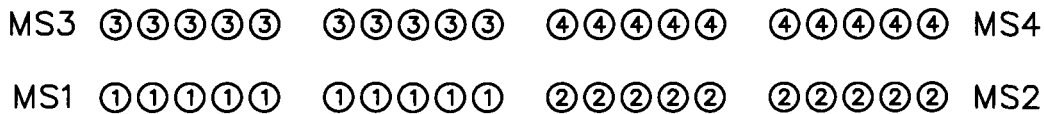
This is a dangerously high particle velocity, two delays were introduced to reduce the vibration level. This divided the shot into two series or parts of 20 holes each, with 5,000 lbs. per delay.



20 Holes Fired Per Delay

$$V = 100 \left( \frac{1000}{\sqrt{5000}} \right)^{-1.6} = 1.44 \text{ in / s}$$

If two more delays MS3 and MS4 were introduced, reducing the number of holes per delay to 10 and the charge per delay to 2,500 lbs., the probable particle velocity can be calculated.



10 Holes Fired Per Delay

$$V = 100 \left( \frac{1000}{\sqrt{2500}} \right)^{-1.6} = 0.83 \text{ in/s}$$

Thus a significant reduction in vibration level can be achieved by the use of delays. Why does delay blasting reduce vibration? The answer is fairly simple, but to understand it one must understand the difference between particle velocity and propagation velocity.

### PROPAGATION VELOCITY VS. PARTICLE VELOCITY

Propagation velocity is more familiar. It is the speed at which a seismic wave travels through the earth from shot to sensor and beyond. The general range of values is from 1,000 to 20,000 ft/s. For a given area, the value is approximately constant.

Particle velocity is quite different. A rock particle vibrates in an elliptical orbit around a rest position. A simple example of particle motion and velocity is the motion of a fisherman in a boat. A passing speedboat generates a wave that passes under the fisherman, causing his boat to oscillate up and down. This is a particle motion. The speed at which it oscillates is particle velocity. Particle velocity is measured in inches per second (in/s) and is the parameter measured by the seismograph.

Delay blasting works or reduces the ground vibration because the seismic wave generated by one delay has traveled a considerable distance due to its propagation velocity before the next delay has fired. The second seismic wave travels at the same propagation velocity as the first and can never catch up to the first. So the seismic waves or vibrations are separated. The following Figure 3 illustrates the process.

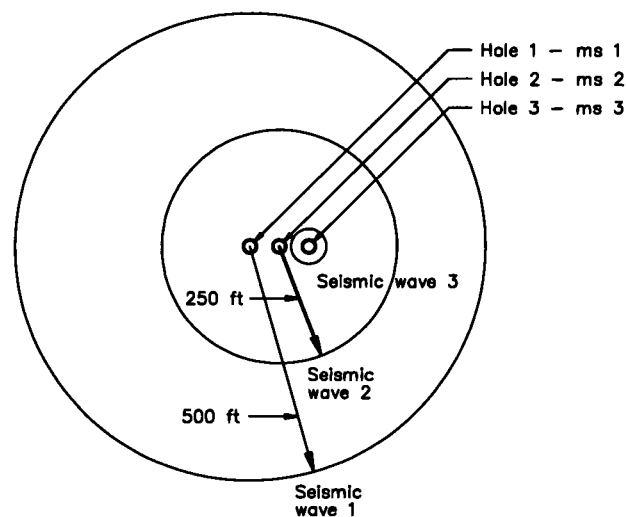


Figure 3. Seismic Waves from Delay Blasting



**PARTICLE VELOCITY - SCALED DISTANCE GRAPH**

This method involves seismic measurement in addition to calculating the scaled distance values from the blast data. Scaled distance is defined as actual distance in feet by the square root of the pounds of explosives fired per delay.

Data is then plotted on log-log graph paper with particle velocity on the vertical axis and scaled distance on the horizontal axis. To be effective, there must be a spread of data from low to high values. This can be accomplished fairly simple by placing the seismograph at increasingly greater distances on successive shots.

Plot the data on the graph, one point for each particle velocity-scaled distance pair. When all the points are plotted, a straight line or envelope should be drawn on the graph so that all the points are below the line. A reasonably accurate eyeball fit is sufficient (Figure 4).

After the data is plotted and the envelope line drawn in, a working value of scaled distance can be read off the graph using this procedure. Start on the particle velocity scale at the specified regulatory particle velocity, e.g., 1.0 in/s. Draw a line horizontally across the graph until it intersects the envelope line. At the point of intersection, drop a vertical line down to the scaled distance axis. The point at which it touches the scaled distance axis is the working value for scaled distance. This value will insure that particle velocities generated by blasting will be less than 1.0 in/s.

If the regulatory value for particle velocity is different from 1.0 in/s, like 2.0 in/s or 0.5 in/s, then start at the proper value and proceed in the same way in Figure 4.

**TABLE 2 VIBRATION DATA**

SHOT	DISTANCE (d)	CHARGE WEIGHT (W)	$\sqrt{W}$	SCALED DISTANCE (Ds)	PARTICLE VELOCITY
1	275	406	20.15	13.65	1.74
2	385	348	18.65	20.64	0.72
3	590	291	17.06	34.59	0.34
4	790	286	16.91	46.71	0.21
5	1060	362	19.03	55.71	0.17

The working value for scaled distance read from the graph is  $D_s = 19$ . This value can now be used to calculate charge weights and distances that will produce vibration levels less than 1.0 in/s.

For either the average method or the particle velocity-scaled distance method, an on-going addition of data as it occurs should be made. The square dot represents a shot that produced an undesirably high particle velocity due to propagation, cap scatter, bad drilling control, overloading the hole or whatever the cause. The high vibration shows up above the envelope line. Thus, the operator can take immediate steps to control the vibration. Also, a safety factor should be added to the adjusted  $D_s$  value. If the adjusted value is 19, then use a value of 23 or 25 as a safety factor.

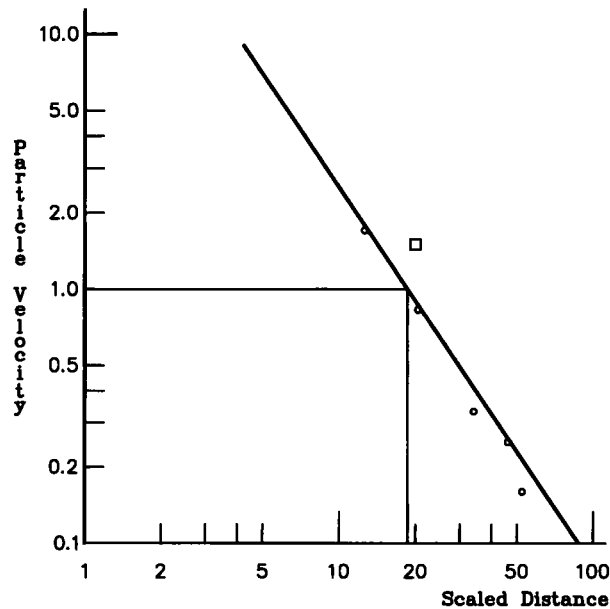


Figure 4. Particle Velocity vs. Scaled Distance

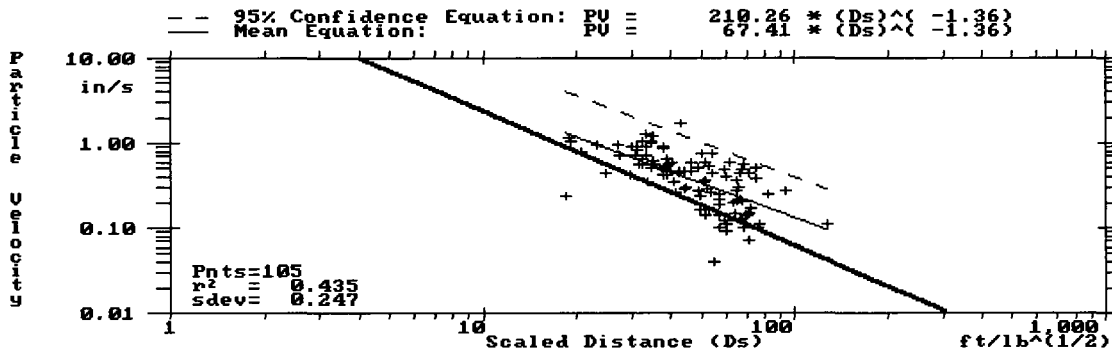
#### GROUND CALIBRATION

Ground calibration should be done when entering a new area or starting a new project. The two principal factors that affect vibration level are charge weight and distance. In addition, rock type, rock density, presence or absence of rock layering, slope of layers, nature of the terrain, blasthole conditions, presence or absence of water, all combine to influence the transmission of vibration. The simplest way to evaluate these factors is by observation of the vibration levels generated. This is called ground or area calibration.

Ground or area calibration can be accomplished by a scaled distance-particle velocity plot on log-log graph paper using data from a series of blasts as discussed previously. A minimum number of five shots will serve as a starter with more data added as additional shots are fired and recorded. The method synthesizes the many factors affecting vibration transmission and enables the operator to determine a safe working value for the scaled distance. Once the scaled distance is adequately determined, all shots should generate vibration levels less than the corresponding particle

**GROUND CALIBRATION IN TRAPROCK (DIABASE)**

Figure 5. is a site specific ground calibration in rock similar to that at Indian Point. The line on the graph is the ground vibration prediction line using the US Bureau formula. The 95% confidence line is the site specific prediction equation generated from actual data. (Appendix 1)



**Figure 5. Site Specific Prediction Equation for Diabase**

The prediction equation from Figure 5. can also be used to predict different charge weights for specific vibration levels at different distances. Table 3. shows the charge weight which can be used at different distances to produce specific amounts of vibration(PPV).

The pounds of explosive which can be fired per delay to generate specific maximum vibration levels (PPV) is shown for difference distances in table 2. These values were calculated using the propagation equation from figure 5.

**TABLE 3. CHARGE WEIGHT /DELAY , PPV AT SPECIFIED DISTANCES**

DISTANCE (FT)	PPV=0.25 IPS	PPV=0.50 IPS	PPV=1.0 IPS	PPV=2.0IPS
330	5.3 lb	14 lb	41 lb	114 lb
960	45 lb	124 lb	346 lb	962 lb
1660	134 lb	372 lb	1035 lb	2880 lb

**TENTATIVE BLASTING PATTERN**

A tentative blast design was prepared to determine the actual amount of explosive which would be used in the blasts. The design consisted of three-inch diameter blastholes drilled on a bench 25 feet high (Appendix 2). An explosive load of 48 pounds would be used per blasthole when blasting at distances greater than 350 ft from protected structures. At distances of less than 350 ft. blastholes can be deck loaded. All holes/charges would be independently delayed.

The anticipated maximum vibration utilizing this blast design for the trenches and the cooling towers would produce a maximum vibration (PPV) at the nuclear reactors as shown in Table 4. Blasting for the cooling towers would be 960 feet from the reactors at their closest point. The maximum anticipated vibration from blasting for the cooling towers is 0.26 inches per second. The 0.26 inches per second is 13% of the safe limit for residential structures. There is no possibility of any damage to any structure at Indian point at these vibration levels.

**TABLE 4. PROJECTED VIBRATION (PPV) AT DIFFERENT DISTANCES**

<b>DISTANCE (FT)</b>	<b>CHARGE WEIGHT (LB)</b>	<b>PPV (IN/SEC)</b>
<b>330</b>	24	0.70
<b>960</b>	48	0.26
<b>1660</b>	48	0.12

**VIBRATION STANDARDS**

The present vibration standards are the result of more than eighty years of research and investigation by concerned scientists. Houses, residential structures, are the weakest type of construction and those of most concern by the average person. Industrial structures are stronger and can tolerate higher vibration levels. Nuclear reactors are built stronger than most industrial structures

Vibration standards for residential structures will be discussed since they are the weakest structure for which mountains of vibration data exist. Vibration standards and limits are placed on residential structures by government agencies which have regulatory authority. The first significant investigation on residential structures was initiated by the U.S. Bureau of Mines in 1930, and culminated in 1942 with publication of Bulletin 442, Seismic Effects of Quarry Blasting. This and other programs will be briefly described.

Thoenen and Windes. Seismic Effects of Quarry Blasting U.S. Bureau of Mines, Bulletin 442, 1942.

<u>Acceleration Index</u>		
Safe zone	-	less than 0.1 g
Caution zone	-	between 0.1 and 1.0 g
Damage zone	-	greater than 1.0 g

Crandell, F. J. Ground Vibration Due to Blasting and Its Effect Upon Structures. Journal of the Boston Society of Civil Engineers, 1949.

$$\text{Energy Ratio Index } ER = \left(\frac{a}{f}\right)^2$$

where:

a = Acceleration (ft/s<sup>2</sup>)  
f = Frequency (Hz)

Safe zone = ER less than 3  
Caution zone = ER between 3 and 6  
Damage zone = ER greater than 6

Energy Ratio has the dimension of velocity and an ER = 1 is equivalent to a particle velocity = 1.9 in/s

Langefors, Westerberg and Kihlstrom. Ground Vibration in Blasting, Parts I-III, Water Power, 1958.

Velocity Index

No damage - less than 2.8 in/s  
Fine cracks - 4.3 in/s  
Cracking - 6.3 in/s  
Serious cracking - 9.1 in/s

Edwards and Northwood. Experimental Blasting Studies on Structures. National Research Council. Ottawa: Canada, 1959.

Velocity Index

Safe zone - Less than 2.0 in/s  
Damage - 4.0 to 5.0 in/s

Nichols, Johnson and Duvall, Blasting Vibration and Their Effects on Structures. U. S. Bureau of Mines, Bulletin, 656, 1971.

Velocity Index

Safe zone - less than 2.0 in/s  
Damage zone - greater than 2.0 in/s

In addition to the Bureau's own work, Bulletin 656 is also a synthesis of the work of the number of other investigators. Particle velocity is considered to be the best measure of damage potential. The safe vibration criterion was specified in Bulletin 656 as follows:

The safe vibration criterion is based on the measurement of individual components, and if the particle velocity of any component exceeds 2 in/s damage is likely to occur.

Damage means the development of fine cracks in plaster. Very quickly the particle velocity, 2 in/s, became known as the Safe Limit. Many regulations were and continue to be still based on this value. Additional levels of vibration based on the results of other investigations used in Bulletin 656 are the following:

Threshold of damage (4 in/s)  
opening of old cracks  
formation of new cracks  
dislodging of loose objects

Minor damage (5.4 in/s)  
fallen plaster  
broken windows  
fine cracks in masonry  
no weakening structure

Major damage (7.6 in/s)  
large cracks in masonry  
shifting of foundation-bearing walls  
serious weakening of structure

The major damage zone correlates reasonably well with the beginning damage level for natural earthquakes.

#### RECENT DAMAGE CRITERIA

In 1980, the U.S. Bureau of Mines reported on its most recent investigation of surface mine blasting in R.I. 8507 (Siskind, et al). Structural resonance responding to low frequency ground vibration, resulting in increased displacement and strain, was found to be a serious problem.

This reintroduced the dependence of damage on frequency. Prior to this, the safe limit particle velocity was independent of frequency. Also, measurements were made inside structures rather than just by ground measurements. Inside measurement seems quite reasonable and logical, but data from previous investigations of structural vibration yielded very poor results, hence, the emphasis on ground measurement.

The threshold of damage used in RI 8507 was specified as cosmetic damage of the most superficial type, of interior cracking that develops in all homes, independent of blasting.

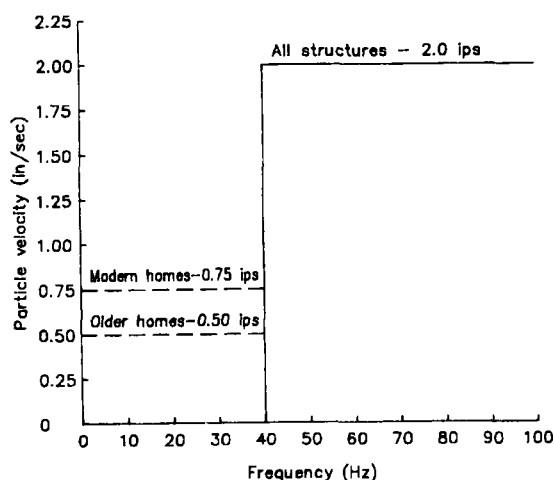
The safe vibration level was defined as levels unlikely to produce interior cracking or other damages in residences.

Safe vibration levels as specified in RI 8507 are given in Table 5. These criteria are based on a 5% probability of damage.

**TABLE 5. SAFE PEAK PARTICLE VELOCITY FOR RESIDENTIAL STRUCTURES  
 (RI 8507)**

TYPE OF STRUCTURE	f < 40 Hz	f > 40 Hz
Modern homes - drywall interiors	0.75 in/s	2 in/s
Older homes - plaster on wood lath for interior walls	0.50 in/s	2 in/s

These safe vibration levels represent a conservative approach to damage and have been the subject of intense criticism by the blasting industry.



**Figure 5. Safe Vibration Levels (RI 8507)**

**ALTERNATIVE BLASTING CRITERIA**

RI 8507 also proposed alternative blasting criteria using a combination of displacement and velocity criteria applied over several frequency ranges. These alternative criteria are shown in Figure 6.

These criteria using both displacement and velocity over respective frequency ranges have not been accepted by all concerned. Instrumentation will need frequency reading capability in addition to the capability of reading both displacement and velocity in order to cover all ranges. This indicates the state of flux in which the question of safe vibration standards existed, which still exists today.

The problem is associated with the concept of what really constitutes vibration damage. The most superficial type of cracking advocated in RI 8507, while not to be condoned, is scarcely a realistic guide for control.

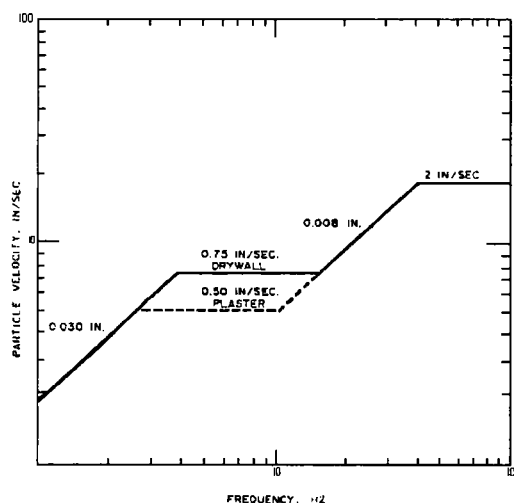


Figure 6. Alternative Blasting Level Criteria Source: RI 8507, U.S. Bureau of Mines

An important consideration to be noted is that there probably is no lower limit beyond which damage will not occur, since there will always be structures at the point of failure due to normal environmental stresses. It is not unusual to read of a structure collapsing for no apparent reason.

In RI 8896, (1984), "Effects of Repeated Blasting on a Wood-Frame House" U.S. Bureau of Mines, it indicates that cosmetic cracks occurred during construction of a test house and also during periods when no blasts were detonated. It was further noticed that human activity, temperature, and humidity changes caused strains equivalent to ground particle velocity of 1.2 in/s to 3.0 in/sec.

#### VIBRATION EFFECTS

Cracks produced in structures by natural earthquakes, which are low intensity effects, have a characteristic pattern called the X - crack or vibration crack. These cracks result from the fact that the top of a structure, due to its inertia, lags behind. The structure is deformed from a regular rectangular shape into a parallelogram, with one of its diagonals elongated and the other compressed. If the elongation exceeds the strength of the material, it will fail producing a tension crack. As the earth vibration reverses, the same thing will occur in reverse, with the opposite diagonals being elongated and compressed with the possible formation of another tension crack. When both cracks occur they form an X - crack pattern. Figure 7. illustrates the process. If it occurs, the X - crack pattern is most likely to be associated with large blasts.



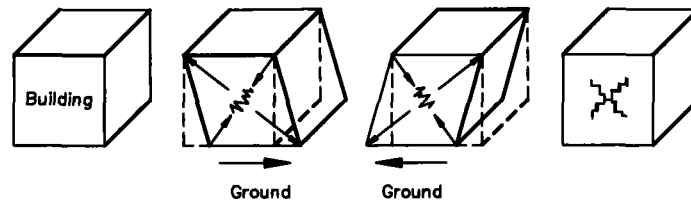


Figure 7. Vibration X - Crack Pattern

#### DIRECTIONAL VIBRATIONAL EFFECTS

The energy that moves out from the source of the blast, measured in terms of ground vibration and peak particle velocity, moves out in all directions from the source. If the ground would transmit vibration in the same manner in all directions and if all other factors remain constant, then theoretically at the same distance in any direction from a blast, the vibration levels would be equal. Unfortunately, on true job conditions, vibration transmission is not ideal and because of changes in the earth structure, vibration is transferred differently in different directions. The geologic structure, joints and faults, will change vibration levels and frequency in different directions of the source. Other factors dealing with blasting pattern design can also contribute to these directional vibration effects.

In the past, it was common practice to monitor behind the blast at the nearest structure since it was assumed that in this direction vibration levels would be greatest. Recommendations for monitoring practice have changed and research has shown that the highest vibration levels are commonly, not behind the shot, but to the sides of the blast. In particular, vibration levels are commonly highest in the direction towards which the delays are progressing. For example, if a blast is fired with the first hole on the left hand side of the pattern and the delays are progressing toward the right hand side of the pattern, then in the direction toward the right hand side of the pattern one would commonly find the highest vibration levels.

In order to calibrate the ground and determine site specific transmission characteristics, it is recommended that at least two seismographs be used when blasting in close proximity to structures. One seismograph placed on the end of the shot and one at 90 degrees. For example, behind the blast. After test shooting is completed and the transmission characteristics are known, the second seismograph may be unnecessary since the ground has already been calibrated and vibration levels in one direction can be related to vibration levels in the other direction.

#### FREQUENCY WAVE LENGTH EFFECTS

When a line of increased motion occurs, what are its dimensions and how large an area is affected? It will cover a space of the order of one to two wavelengths. Wavelength is defined as propagation velocity multiplied by the wave period

$$L = V T$$

where:

- L = Wavelength (ft)
- V = Propagation velocity (ft/s)
- T = Wave period (s)

For a wave of period 1/10 sec and propagation velocity 2,000 ft/s, the wavelength is 200 feet.

Assuming the waves are approximately the same (Fig. 8.), at maximum coincidence the motion would be doubled but the wave length will be that of either wave since they are the same (Figure 9.).

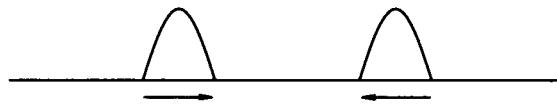


Figure 8. Converging Equal Wavelets

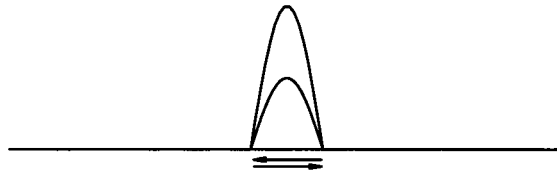


Figure 9. Composite Wave Motion at Maximum Coincidence

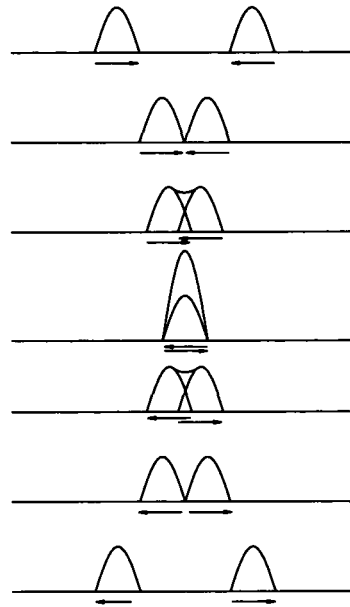
This form will be repeated after the maximum has occurred when the waves pass complete coincidence and begin to separate each into its own distinct form. Thus, there is a periodicity whose wavelength approaches the sum of the two wavelengths. Also, the wavelength of the composite motion varies from a single wavelength to approximately double the single wavelength. The converging and diverging wavelets are shown in Figure 10. and the resulting composite motion is shown in Figure 11.

The wave period and the frequency are both effected. At the point of maximum coincidence the period and frequency are those of the single wave. Since the period may approach double that of a single wave, the frequency will be cut approximately in half.

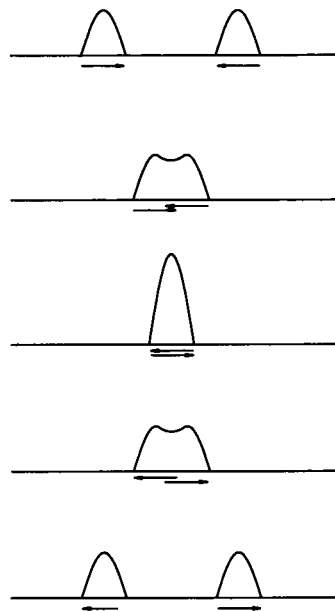
The significant points here are that they can exist.

1. A region of increased seismic motion and hence increased peak particle velocity with maximum at the center, minimum at the edges of the resultant combined waves.
2. The region in which this occurs, the order of two wave lengths wide approximately 400 to 800 feet depending on propagation velocity and wave period.

3. Wave periods will be increased to approximately double with a corresponding lowering of the frequency to half.
4. A region of high-seismic risk because of the increased motion and reduced frequency of vibration.



**Figure 10. Converging and Diverging Wave Interaction**



**Figure 11. Composite Motion**

## NON-DAMAGE EFFECTS

Damage producing vibration seldom occurs, but many other effects occur that are disconcerting and alarming to persons who feel and hear the vibration. Some of these effects are:

- Walls and floors vibrate and make noise.
- Pipes and duct work may rattle.
- Loose objects, plates, etc., may rattle.
- Objects may slide over a table or shelf, and may fall off.
- Chandeliers and hanging objects may swing.
- Water may ripple and oscillate.
- Noise inside a structure is greatly amplified over noise outside.
- Vibration is very disturbing to occupants.

## CAUSES FOR CRACKS OTHER THAN BLASTING

Cracking is a normal occurrence in the walls and ceilings of structures, and the causes are multiple, ranging from poor construction to normal environmental stress, such as thermal stresses, wind, etc. The Small Home, published by the Architects Small House Service Bureau of the United States, Inc. 1925, gave a list of reasons for the development of cracks, which included the following:

- Building a house on a hill.
- Failure to make the footings wide enough.
- Failure to carry the footings below the frost line.
- Width of footings not made proportional to the loads they carry.
- The posts in the basement not provided with separate footings.
- Failure to provide a base raised above the basement floor line for the setting of wooden posts.
- Not enough cement used in the concrete.
- Dirty sand or gravel used in the concrete.
- Failure to protect beams and sills from rotting through dampness.
- Setting floor joists one end on masonry and the other end on wood.
- Wooden beams used to support masonry over openings.
- Mortar, plaster, or concrete work allowed to freeze before setting.
- Braces omitted in wooden walls.
- Sheathing omitted in wooden walls (excepting in "back-plastered" construction).
- Drainage water from roof not carried away from foundations.
- Floor joists not bridged.
- Supporting posts too small.
- Cross beams too light.
- Sub-flooring omitted.
- Wooden walls not framed so as to equalize shrinkage.
- Poor materials used in plaster.
- Plaster applied too thin.

- Lath placed to close together.
- Lath run behind studs at corners.
- Metal reinforcement omitted in plaster at corners.
- Metal lath omitted where wooden walls join masonry.
- Metal lath omitted on wide expanses of ceiling.
- Plaster applied directly on masonry at chimney stack.
- Plaster applied on lath that is too dry.
- Too much cement in the stucco.
- Stucco not kept wet until set.
- Subsoil drainage not carried away from walls.
- First coat of plaster not properly keyed to backing.
- Floor joists placed too far apart.
- Wood beams spanned too long between posts.
- Failure to use double joists under unsupported partitions.
- Too few nails used.
- Rafters too light or too far apart.
- Failure to erect trusses over wide wooden openings.

\* Published in Monthly Service Bulletin 44 of the Architects' Small House Service Bureau of the United States, Inc.

#### BLAST DESIGN TO REDUCE VIBRATION LEVELS

When vibration levels are too high and it becomes desirable and even necessary to reduce them, there are a number of options.

#### CHARGE REDUCTION

The maximum charge per delay may be reduced by decreasing the number of holes per delay. If the number of holes per delay cannot be reduced then it may be possible to deck load and fire each hole with two or more delays.

#### BLAST DESIGN

The vibration level can be reduced by redesigning the blast so that less energy per hole is necessary to fragment the rock. This may require changing the hole spacing, the burden and even the hole size. A change in explosive may be helpful also. This requires going back to square one and starting over. This is an extreme circumstance and not likely to be necessary.

#### BLASTING STANDARD FOR NON-RESIDENTIAL STRUCTURES

Vibration standards can be divided into two other groups in addition to the normal building standards, high level vibration structures and low level vibration sensitive components.

#### BLASTING NEAR CONCRETE STRUCTURES

On many demolition projects, old concrete is near the blasting operation. In fact, it is not uncommon to blast away part of a structure, leaving the other structure intact. This is a common procedure when locks along rivers need to be refurbished. When locks become eroded due to the water and the environmental conditions, approximately two feet of old concrete is blasted away and new concrete is poured in its place. It is obvious that the concrete that remains from the original structure has been subjected to very high peak particle velocity. Oriard measured values of strain and peak particle velocity that produced various types of failure in concrete. His results are given in Table 6.

**TABLE 6. FAILURE IN CONCRETE DUE TO VIBRATION**

<b>TYPE</b>	<b>STRAIN (<math>\mu\text{in/in}</math>)</b>	<b>PPV (in/s)</b>
Static	140	20
Grout Spall	700	100
Skin Spall	1300	200
Cracking	2400	375

#### GREEN CONCRETE

Concrete and bridges fall into the high level vibration structures. Green concrete, however, is not in this group. Different types of concrete exist. Therefore, general conservative guidelines for concrete may be given. Since concrete acquires about one third its strength in 72 hours, after this time a peak particle velocity of 1.0 in/s is a reasonable maximum until the concrete reaches full strength at 28 days. Before 72 hours it is not advisable to blast.

#### BLASTING NEAR GREEN CONCRETE

It is not uncommon to have blasting operations in one section of a project and the pouring of concrete in another. Contractors do have concern as to what effect the blasting vibration has on the integrity of the new structure being poured. Some guidelines for peak particle velocities related to time after pouring are given in Table 7.

**TABLE 7. VIBRATION LEVELS FOR GREEN CONCRETE**

<b>TIME AFTER POUR (HOURS)</b>	<b>PPV (in/s)</b>
0 - 4 Hours	2.00

4 - 24	Hours	0.25
1 - 3	Days	1.00
3 - 7	Days	2.00
7 - 10	Days	5.00
> 10	Days	10.00

### SENSITIVITY TO VIBRATION

Human beings are remarkably sensitive to vibration. If this were not so, the vibration problem would scarcely exist. The explosive technology of today insures that most operations are conducted in a safe manner. In relatively few cases is there a significant probability of damage.

Since vibration is felt in practically all cases, the reaction to this sensation is one of curiosity, concern, and even fear. Hence, it is important to understand something about human response to vibration that depends on vibration levels, frequency and duration. In addition to these physical factors, it is important to keep in mind that human response is a highly subjective phenomenon.

Human response has been investigated by many researchers. One of the early investigations was by Reiher and Meister, Berlin, 1931. Other investigations were made by Goldman, 1948, and Wiss and Parmelee, 1974. A composite of these investigators' results was presented graphically in the U. S. Bureau of Mines RI 8507, Siskind, et al, 1980. This composite is represented here in Figure 12.

The human response curves are all similar and highly subjective in that the response is a mixture of physiological and psychological factors individual to each person. Based on these curves, a very simple and practical set of human responses can be designated as shown in Figure 8.

**TABLE 8. HUMAN RESPONSE**

RESPONSE	PARTICLE VELOCITY	DISPLACEMENT AT 10 Hz	DISPLACEMENT AT 40 Hz
Noticeable	0.02 in/s	0.00032 in	0.00008 in
Troublesome	0.2 in/s	0.0032 in	0.0008 in
Severe	0.7 in/s	0.011 in	0.0028 in

Vibration is a fact of daily life, which one regularly experiences but is seldom aware of. This type of vibration has been designated cultural vibration. Generally, it elicits no reaction from the person affected.

Other vibration that contrasts sharply, because it is not part of the daily experience but is unusual, has been designated A-cultural. It surprises a person, is disturbing, and causes an acute awareness.

Some examples of cultural and A-cultural vibration are listed in the following:

**CULTURAL VIBRATION**

Automobile  
 Commuter Train  
 Household  
 Industrial Plant or Office  
 Airplane

**Common Denominator:**

No reaction

**A-CULTURAL VIBRATION**

Blasting  
 Pile Driving  
 Impact Machinery  
 Jack Hammer  
 Forging Hammers

**Common Denominator:**

Persons react because these vibrations  
 are unfamiliar, disturbing

Blasting is definitely A-cultural for the average person. The annoyance and fear associated with it begin at levels much lower than the damage level for structures.

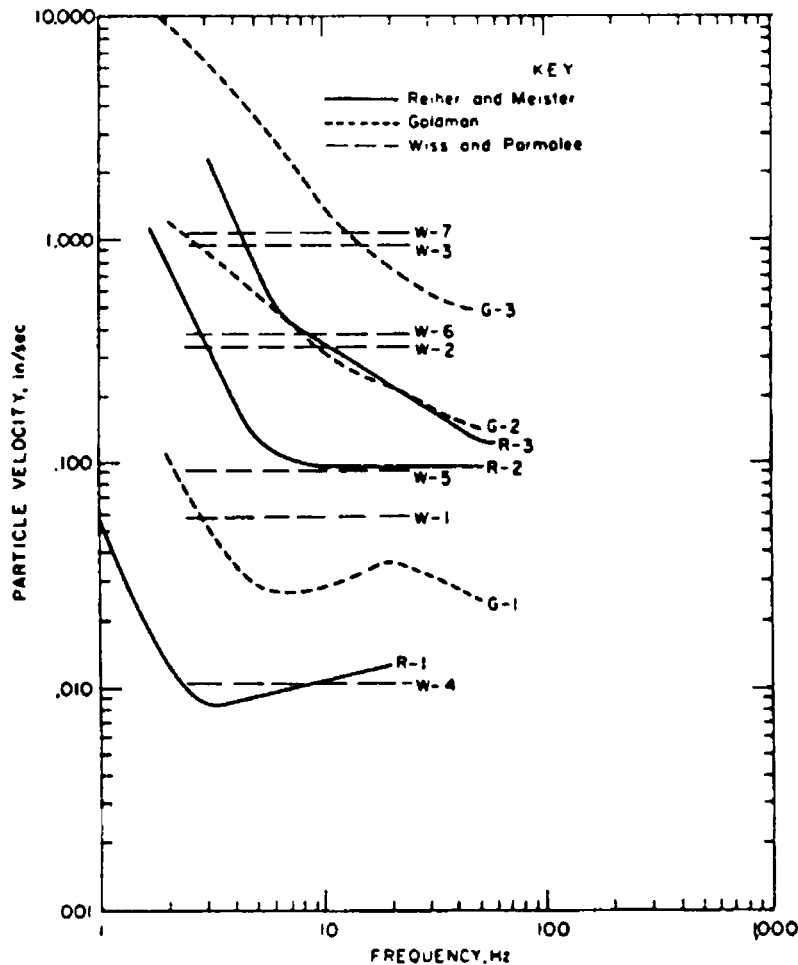


Figure 12. Human Response To Vibration (RI 8507)



ENVIRONMENTAL VIBRATION

Blast vibrations are sensed by individuals at very low levels. Blasting vibration is A-cultural vibration and because the public equates blasting and explosives with death and destruction rather than progress and improvements in quality of life they are apprehensive about any blasting vibration that they sense.

All activities produce some amount of vibration and are constantly present in homes and structures. Environmental factors such as wind, heating and cooling, changes in humidity, traffic, trains, thunder, fireworks and minor earthquakes all produce stress in a structure. The research conducted by the United States Bureau of Mines showed that strains equivalent to those produced by blast vibrations of three inches per second could result from normal environmental stresses. In most cases the public are either unaware or not concerned by the effects of environmental vibration.

ACTIVITY	SCALE																			
	1	2	3	4	5	6	7	8	9	10	11	12	13	14	15	16	17	18	19	20
WALKING	X	X	X	X	X															
TRAIN NEARBY	X	X	X	X	X	X														
WALKING ON WOOD FLOOR	X	X	X	X	X	X	X													
PILE DRIVING, PUNCH BARGE	X	X	X	X	X	X	X	X												
GARBAGE DISPOSAL	X	X	X	X	X	X	X	X	X											
JUMPING	X	X	X	X	X	X	X	X	X											
DOOR SLAMS	X	X	X	X	X	X	X	X	X											
POUNDING NAILS	X	X	X	X	X	X	X	X	X	X										
DAILY ENVIRONMENTAL CHANGE	X	X	X	X	X	X	X	X	X	X	X									
RIDING IN AUTOMOBILE	X	X	X	X	X	X	X	X	X	X	X	X	X							
PEAK PARTICLE VELOCITY	0.001	0.002	0.004	0.008	0.016	0.032	0.064	0.128	0.256	0.512	1.024	2.048	4.096	8.192	16.38	32.77	65.54	131.07	262.14	524.29

Figure 13. Konya's Environmental Vibration Scale™

KONYA'S VIBRATION SCALE

It is often difficult for the public to understand the magnitudes of vibration from blasting and relate this to normal environmental vibration which they sense every day. Since blast vibration is A-cultural and triggers response people become concerned about vibration levels from blasting while they are not concerned about the same vibration levels from cultural vibration which occurs every day in their lives. To put vibration in the proper perspective we can compare both the A-cultural and cultural vibration magnitudes. To do this in a simple understandable manner use the Konya Scale where we can divide the vibration levels into 20 different classes. We can start with a peak particle velocity of 0.001 to less than 0.002 inches per second and put all vibration less than 0.002 in class one. Class one is the level at which some (very few) people can perceive vibration. We then double the previous number from 0.001 to 0.002. Class two vibration would be 0.002 to less than 0.004. Class three would double again to 0.004 but less than 0.008 and so on.

This class method can be used for both blast effects and separately for environmental vibration. The two charts can then be easily compared without confusion. Konya's Blast Effects

Scale shows the PPV levels and the class numbers for Human perception and potential damage which can result at high vibration levels. Konya's Environmental Vibration Scale shows vibration levels from normal activities (Figure 13.).

For example, class five vibration is the level where most people perceive vibration (Konya's Blast Effects Scale (Figure 14.)) and some become concerned that the vibration will damage their home. Class five on Konya's Environmental Vibration Scale shows that all normal activities on the chart produce vibration at class five or greater. In general most regulatory bodies allow vibration to at least class 10 because they understand that no structural damage can occur in homes at these vibration levels.

EFFECTS	VIBRATION CLASS NUMBER																			
	1	2	3	4	5	6	7	8	9	10	11	12	13	14	15	16	17	18	19	20
PERCEPTION BY OLDER POPULATION	X	X	X	X	X	X	X	X	X	X	X	X	X	X	X	X	X	X	X	X
PERCEPTION BY ALL					X	X	X	X	X	X	X	X	X	X	X	X	X	X	X	X
WATER RIPPLES						X	X	X	X	X	X	X	X	X	X	X	X	X	X	X
PIPES RATTLE					X	X	X	X	X	X	X	X	X	X	X	X	X	X	X	X
LOOSE OBJECTS RATTLE					X	X	X	X	X	X	X	X	X	X	X	X	X	X	X	X
CRACK EXTENSIONS IN PLASTER (INVISIBLE)											X	X	X	X	X	X	X	X	X	X
CRACK EXTENSIONS (VISIBLE)												X	X	X	X	X	X	X	X	X
NEW CRACK FORMATION (PLASTER)												X	X	X	X	X	X	X	X	X
FINE CRACKS IN MASONRY													X	X	X	X	X	X	X	X
BROKEN WINDOWS													X	X	X	X	X	X	X	X
CHIMNEY DAMAGE													X	X	X	X	X	X	X	X
LARGE CRACKS IN MASONRY WALLS														X	X	X	X	X	X	X
CRACKS IN CONCRETE WALLS																		X	X	X
CRACKS IN CONCRETE SLABS																			X	X
CRACKS IN MASSIVE CONCRETE																				X
PEAK PARTICLE VELOCITY	0.001	0.002	0.004	0.008	0.016	0.032	0.064	0.128	0.256	0.512	1.024	2.048	4.096	8.192	16.38	32.77	65.54	131.07	262.14	524.29
IN INCHES/SECOND																				
(THRESHOLD VALUES)																				

Figure 14. Konya's Blast Effects Scale <sup>TM</sup>

COMPARISON OF BLAST AND EARTHQUAKE VIBRATION

An earthquake is a regional event, while blasting is a local event. Blasting produces a vibration which rapidly decays in intensity in distances measured in feet while the decay of vibration from earthquakes is measured in miles from the source.

To understand the differences between earthquake and blasting magnitude one can compare the same magnitude particle velocity for earthquake frequencies and that from blast vibrations in geologic materials at Indian Point. Earthquake frequencies are generally less than one hertz while blasting frequencies for close proximity construction blasting is above 50 hertz. At the same measured particle velocity the movement (displacement) of the earth is greater in earthquakes than in blasting. Table 9. shows a comparison in displacement for the same particle velocities at different frequencies.

A Mercalli Scale intensity of VII would produce a maximum peak particle velocity of about 6.3 inches per second. At a one hertz earthquake frequency this would produce a displacement of 1.114 inches, while a blast vibration producing the same peak particle velocity of 6.3 inches per second at a frequency of 50 hertz would produce a displacement of 0.020 inches. The earthquake ground displacement would be 315 times greater than that of blasting at the same particle velocity.

**TABLE 9. COMPARISON OF DISPLACEMENT IN INCHES**

MERCALLI MAGNITUDE	RICHTER SCALE	PPV at 1 Hz, (in)	DISP at 1 Hz, (in)	DISP at 50 Hz, (in)
I	-1 to 3.5	0.002 to 0.35	0.0003 to 0.0557	6.37E-06 to 0.0011
II	3.5 to 5.4	0.35 to 3.5	0.0557 to 0.5570	0.0011 to 0.0111
III	3.5 to 5.4	0.35 to 3.5	0.0557 to 0.5570	0.0011 to 0.0111
IV	3.5 to 5.4	0.35 to 3.5	0.0557 to 0.5570	0.0011 to 0.0111
V	3.5 to 5.4	0.35 to 3.5	0.0557 to 0.5570	0.0011 to 0.0111
VI	3.5 to 5.4	0.35 to 3.5	0.0557 to 0.5570	0.0011 to 0.0111
VII	5.4 to 6.0	3.5 to 6.3	0.5570 to 1.003	0.0111 to 0.0200
VIII	6.1 to 6.9	7.0 to 17.7	1.114 to 2.810	0.0222 to 0.0563
IX	7.0 to 7.9	19.9 to 56	3.168 to 8.917	0.0634 to 0.1780
X	Over 8.0	Above 62.8	Above 10.000	Above 0.2000
XI	Over 8.0	Above 62.8	Above 10.000	Above 0.2000
XII	Over 8.0	Above 62.8	Above 10.000	Above 0.2000

**PROPOSED SAFE BLASTING VIBRATION LIMITS**

The review conducted of the past earthquake vibration standards at Indian Point, standards for weakly constructed residential structures, industrial structures, and 80 years of research data proves that blasting can be safely conducted at the reactor without any damage to the structure.

An extremely conservative vibration limit would be 2.0 inches per second which is the limit for residential structures. This vibration limit would protect the structures and allow construction at reasonable costs. According to the available data the reactor should be totally safe at **three times** this vibration limit if in fact it is built to withstand a Mercalli Intensity VII earthquake.

It is obvious that blasting mats must be used for cover of the blasting area to protect the electrical wires and any other delicate structure which could be damaged by small pieces of flyrock.

The excavation for the tower will vary in distance from 960 feet to 1660 feet from the reactors.

The major trenching would be 330 feet at it's closest point to the reactors. The turbine generator buildings would be about 55 feet from the trench blasting

## ASSESSMENT OF ROCK QUANTITIES AND COSTS

### ROCK TRENCHES

The following measurements were taken from the drawings for the large diameter pipe installations from the cooling towers. Blasting of trench rock costs \$50/cubic yard. The \$50/yd<sup>3</sup> cost would cover the cost of all types of trench blasting on the project

- Four pipe trench, 62 feet wide and 20 feet deep and 1722 feet long equals 79,084 yd<sup>3</sup>
- Two pipe trench, 40 feet wide and 20 feet deep and 2500 feet long equals 74,074 yd<sup>3</sup>
- Total trench rock equals 153,158 yd<sup>3</sup>

At a cost of \$50.00 per yd<sup>3</sup> the total cost of trench blasting would be \$7,657,900

### COOLING TOWERS

The cost of the excavation for the two cooling towers would be \$7/cubic yard.

The volume of rock needed for excavation for the towers would be the total rock excavation minus the rock excavated for the trenches. The total volume was 2,100,000 yd<sup>3</sup> minus the 153,158 cubic yards for trenches leaves 1,946,842 yd<sup>3</sup> for the two towers.

The cost of the excavation for the towers is 1,946,842 yd<sup>3</sup> times \$7.00 yd<sup>3</sup> equals \$13,627,894 (Appendix 3).

### TOTAL DRILLING AND BLASTING COST

The total cost of drilling and blasting would include the drilling and blasting cost, blasting consultant and seismic monitoring cost. These costs are listed below.

- Drilling and blasting = \$7,657,900 + \$13,627,894 = \$21,285,794
- Blasting Consultant = \$500,000
- Seismic Monitoring Cost = \$ 600,000

**The Total Cost of Drilling and Blasting = \$22,385,794**

DISPOSAL OF BROKEN ROCK

There are many different options for the disposal of the 2,100,000 yd<sup>3</sup> (bank) of rock which will be blasted. The rock can be placed on site, dumped into the ocean, sold to quarry or crushed and sold from the site.

For the purpose of this study the ocean dumping cost is the highest and the only sure method of disposal since the rock quality is unknown at this time.

WASTED ON SITE

The most economic method of disposal would be on site. The cost for site disposal would be the cost of loading the material into trucks and the short haul distance with the trucks.

OCEAN DUMPING

The state of New York and the State of New Jersey both are building artificial reefs of their shores. The offshore reefs are used for dumping rock from barges in the current harbor deepening projects. A permit would be needed from the Department of Environmental Conservation to ocean dump rock from the Indian Point excavation. This should not be a problem because the DEC encourages the building of these reefs because they attract fish.

Great Lakes Dredge and Dock Company was contacted and provided the cost of moving the rock from Indian Point to an artificial reef. The haul distance was 200 miles round trip. Two 4000 yd<sup>3</sup> barges which hold 4500 tons would be provided as well as an ocean going tug boat. The blasted rock would be loaded into the barge by the construction company at Indian Point. The cost of digging the broken rock and the cost of moving the rock to the barge is not in the price given below. The cost to dump the 2,100,000 yd<sup>3</sup> of rock would be \$18.90 per yd<sup>3</sup>.

If all the material is dumped into the ocean the **cost would be \$39,690,000.**

SALE TO QUARRY

I have talked to Tilcon Quarry Management concerning stone produced from blasting at Indian Point. If the stone is of good quality they may be interested in some business arrangement which could reduce the disposal cost from that of ocean dumping. In my opinion if the stone quality is good the cost of disposal may be cut to about \$4,000,000 to \$6,000,000.

### CRUSHING ON SITE

If the rock is of sufficient quality a crushing contractor could set up a crushing plant on the property and crush the rock for \$5 to \$7 /ton. This crushed rock could possibly be sold to a quarry or contractors and also could be used for the needs of the site construction. If the rock could be sold from the site a profit of at least \$2.00 per ton should be realized. This could produce a revenue of \$5,000,000 to \$10,000,000 for the 4,950,000 tons excavated depending on the quality of the stone.

The minimum quantity of crushed stone necessary for base material under the cooling towers as well as fill around the pipes was estimated as follows. The base under the 700 foot diameter excavations for the two cooling towers would require 28,500 yd<sup>3</sup> or 55, 800 tons of rock base material for a one foot deep base.

Bedding material will be needed surrounding the 12 foot diameter and other pipes. The bedding at minimum would be 103,000 yd<sup>3</sup> or 195,000 tons of fill around pipes.

The 131,500 tons of rock brought into the site would cost about \$15 per ton or about \$29 per cubic yard of crushed stone delivered to the property. This would be at minimum an additional cost of \$4,000,000.

### TOTAL COST OF DRILLING, BLASTING, AND DISPOSAL (LOADING AND HAULAGE COSTS NOT INCLUDED)

The total cost of drilling, blasting, and disposal would be as follows:

<input type="checkbox"/> Drilling and blasting	\$22,385,794
<input type="checkbox"/> Ocean dumping	\$39,690,000.
<input type="checkbox"/> <b>Total cost</b>	<b>\$62,075,794</b>

### EXECUTION OF PROJECT

The project would require a competent drilling and blasting contractor. The contractor should not be selected on low bid alone but also on reputation and past history of successfully completing delicate projects. The contractor must be required to hire a reputable blasting consultant who must be present on the project. The costs shown in this report are conservative and provides for the selection of a competent contractor and blasting consultant.

Core drilling must be done prior to any construction on the project. RQD and percent recovery are important information which must be given the contractors/owners Blasting consultant.

Test blast would be conducted at the farthest distance (1660 feet) from the reactor using small charges to test the scaling factors and determine the site specific vibration decay factors. At least six seismographs would be strategically placed during the test blast phase of the project.

At least four seismographs would be strategically placed and monitored during all production blasting.

#### CONCLUSIONS

It is feasible and safe to blast the rock for the cooling towers and trenches at Indian Point. The Mercalli Intensity of VII for which the facility was designed was the limiting vibration level. It was determined that the blasting could be accomplished at a much lower level or a fraction of the vibration limit. It is suggested that vibration can easily be controlled well below the safe maximum level and that the residential structure vibration limit of 2 inches per second be adopted for all blasting at the facility. The industrial buildings can tolerate much greater vibration limits than residential structures. The actual vibration produced from blasting the cooling towers will be 6% to 13% of safe limits for residential structures and be totally safe for the reactors and other buildings on site. The vibration from the trench blasting will also be well below the residential safe limits.

Cost calculations were conducted for drilling and blasting as well as the rock disposal. The only method of rock disposal that is certain and known at this time is ocean dumping of the waste rock. This is also the most expensive option. Other less expensive methods of rock disposal may be possible once the quality of the rock is known.



APPENDIX 1 – VIBRATION CONSULTANT PROGRAM

Precision Blasting  
 Services, Inc.

VIBRATION CONSULTANT

V5.00 Page: 1  
 Date: 05-20-2003

Number of records: 105 File: PPTLTKM.VIB 2003-Jan-15 Wed 11:02 9,640  
 INDIAN POINT

GROUND VIBRATION DATA (Data points: 105):  
 95% confidence level equation:  $PV = 210.26 * (Ds) ^ (-1.36)$   
 Coefficient of determination ( $r^2$  - "goodness of fit") = 0.435 Standard deviation = 0.247

No.	SHOT	LOCATION	DATE	DAY	TIME	P.P.V.	FREQUENCY	AIR	DISTANCE	CHARGE	SCALED DISTANCES		
							Hz	PRESSURE	ft	WEIGHT	VIBRAT.	AIRPRES.	
						in/s				lb			
1	10-02	IKON	Bot lev wst face	01-23-2002	Wed	10:32	0.94	50	126.00	180.00	60.00	23.24	45.98
2	11-02	IKON	2 lev west face	01-25-2002	Fri	10:46	0.26	100	112.00	325.00	62.00	41.28	82.11
3	12-02	IKON	2 lev west face	01-28-2002	Mon	10:02	0.52	62	118.00	276.00	63.00	34.77	69.36
4	13-02	IKON	2 lev North face	01-30-2002	Wed	09:58	0.35	36	120.00	310.00	83.50	33.92	70.93
5	14-02	IKON	BOF LVL EST FACE	02-05-2002	Tue	09:50	0.43	36	114.00	170.00	19.00	39.00	63.71
6	15-02	IKON	2ND LVL WEST FC	02-26-2002	Tue	10:31	0.56	50	124.00	310.00	76.00	35.56	73.19
7	16-02	IKON	2ND LVL WEST FC	02-27-2002	Wed	10:46	0.30	71	118.00	370.00	68.00	44.87	90.65
8	17-02	IKON	3RD LVL SOUTH FC	03-01-2002	Fri	11:48	0.40	33	116.00	360.00	36.00	60.00	109.03
9	18-02	IKON	B LVL S FC NE PT	03-05-2002	Tue	10:52	0.56	45	117.00	215.00	39.00	34.43	63.40
10	19-02	IKON	2ND LEVEL W	03-07-2002	Thu	09:33	0.92	45	123.00	270.00	81.00	30.00	62.40
11	20-02	IKON	BTM LVL SOUTH FC	03-12-2002	Tue	10:31	0.74	22	121.00	475.00	89.00	50.35	106.39
12	21-02	IKON	2ND LVL WEST FC	03-13-2002	Wed	09:22	0.34	62	120.00	370.00	81.00	41.11	85.51
13	22-02	IKON	BTM LVL STH FACE	03-14-2002	Thu	11:13	0.45	33	112.00	370.00	31.00	66.45	117.78
14	23-02	IKON	BTM LVL WEST FC	03-15-2002	Fri	03:19	0.52	33	114.00	475.00	40.00	75.10	138.89
15	24-02	IKON	BTM LVL WEST FC	03-19-2002	Tue	01:16	0.15	37	110.00	470.00	43.00	71.67	134.16
16	26-02	IKON	BTM LVL WEST FC	03-22-2002	Fri	03:30	0.44	39	116.00	470.00	43.00	71.67	134.16
17	28-02	IKON	BTM LVL WEST FC	03-27-2002	Wed	11:55	0.15	45	112.00	422.00	43.00	64.35	120.45
18	30-02	IKON	BTM LVL WEST FC	04-03-2002	Wed	12:38	0.55	36	116.00	420.00	38.00	68.13	124.93
19	30-02	IKON	BTM LVL WEST FC	04-03-2002	Wed	12:38	0.10	27	110.00	420.00	38.00	68.13	124.93
20	31-02	IKON	BTM LVL SOUTH FC	04-04-2002	Thu	01:26	0.13	50	106.00	430.00	40.00	67.99	125.73
21	32-02	IKON	2ND LVL WEST FC	04-05-2002	Fri	11:19	0.91	50	118.00	197.00	40.00	31.15	57.60
22	32-02	IKON	2ND LVL WEST FC	04-05-2002	Fri	11:19	0.91	50	118.00	197.00	40.00	31.15	57.60
23	33-02	IKON	BTM LVL WEST FC	04-08-2002	Mon	10:40	0.58	45	117.00	382.00	38.00	61.97	113.62
24	34-02	IKON	BTM LVL WEST FC	04-09-2002	Tue	03:38	0.12	63	109.00	420.00	40.00	66.41	122.81
25	35-02	IKON	BTM LVL WEST FC	04-11-2002	Thu	11:16	0.74	31	117.00	345.00	40.00	54.55	100.88
26	36-02	IKON	BTM LEVEL SOUTH	04-12-2002	Fri	11:29	0.25	28	111.00	520.00	40.00	82.22	152.05
27	37-02	IKON	BTM LVL WEST FC	04-15-2002	Mon	12:56	0.25	28	110.00	375.00	43.00	57.19	107.04
28	37-02	IKON	BTM LVL WEST FC	04-15-2002	Mon	12:56	0.14	83	112.00	375.00	43.00	57.19	107.04
29	38-02	IKON	BTM LVL WEST FC	04-17-2002	Wed	11:23	0.37	33	116.00	428.00	43.00	65.27	122.17
30	39-02	IKON	BTM LVL WEST FC	04-18-2002	Thu	10:27	0.44	42	116.00	342.00	40.00	54.07	100.00
31	41-02	IKON	BTM LVL WEST FC	04-23-2002	Tue	12:13	0.49	31	119.00	427.00	40.00	67.51	124.86
32	42-02	IKON	BTM LVL WEST FC	04-24-2002	Wed	10:54	0.22	56	109.00	422.00	40.00	66.72	123.39
33	45-02	IKON	BTM LVL WEST FC	04-29-2002	Mon	12:39	0.09	36	109.00	377.00	39.00	60.37	111.17
34	46-02	IKON	3RD LVL WEST FC	05-01-2002	Wed	12:27	0.10	22	109.00	435.00	40.00	68.78	127.19
35	49-02	IKON	2ND LVL WEST FC	05-06-2002	Mon	10:28	0.56	56	117.00	190.00	36.00	31.67	57.54
36	52-02	IKON	2ND LVL WEST FC	05-08-2002	Wed	12:29	1.04	710	121.00	204.00	40.00	32.26	59.65
37	53-02	IKON	2ND LVL SOUTH FC	05-10-2002	Fri	11:27	0.16	830	112.00	280.00	28.00	52.92	92.21
38	58-02	IKON	2ND LVL WEST FC	05-21-2002	Tue	12:38	0.16	83	109.00	296.00	36.00	49.33	89.64
39	60-02	IKON	2 LVL W FC RAMP	05-23-2002	Thu	02:35	0.96	63	124.00	162.00	36.00	27.00	49.06
40	61-02	IKON	2ND LVL WEST FC	05-28-2002	Tue	12:40	0.29	63	120.00	325.00	54.00	44.23	85.98
41	62-02	IKON	2ND LVL SOUTH FC	05-29-2002	Wed	11:12	0.43	71	116.00	225.00	58.00	29.54	58.13
42	63-02	IKON	2 LVL W FC RAMP	05-31-2002	Fri	01:49	0.44	38	116.00	148.00	36.00	24.67	44.82
43	65-02	IKON	2ND LVL WEST FC	06-03-2002	Mon	03:25	0.57	630	117.00	317.00	65.00	39.32	78.84

Precision Blasting  
 Services, Inc.

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V5.00 Page: 2  
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 INDIAN POINT

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 Coefficient of determination ( $r^2$  - "goodness of fit") = 0.435 Standard deviation = 0.247

No.	SHOT	LOCATION	DATE	DAY	TIME	P.P.V.	FREQUENCY	AIR	DISTANCE	CHARGE	SCALED DISTANCES	
							in/s	Hz	PRESSURE	ft	WEIGHT	VIBRAT.
								dB		lb		
44	64-02	IKON 2ND LVL WEST PC	06-03-2002	Mon	11:07	0.10	630	126.00	475.00	69.00	57.18	115.81
45	66-02	IKON 2ND LVL WEST PC	06-05-2002	Wed	11:14	0.54	56	123.00	475.00	80.00	53.11	110.24
46	67-02	IKON 2ND LVL SOUTH PC	06-06-2002	Thu	10:48	0.73	100	119.00	260.00	72.00	30.64	62.50
47	68-02	IKON 2ND LVL WEST PC	06-10-2002	Mon	11:18	0.51	50	123.00	424.00	75.00	48.96	100.54
48	69-02	IKON 2ND LVL EAST PC	06-12-2002	Wed	01:45	0.64	100	118.00	269.00	68.00	32.62	65.90
49	70-02	IKON 2 LV RAMP W FACE	06-14-2002	Fri	12:27	0.71	28	123.00	217.00	43.00	33.09	61.94
50	71-02	IKON 2ND LVL EAST PC	06-18-2002	Tue	01:54	0.53	71	118.00	307.00	64.00	38.38	76.75
51	72-02	IKON 2ND LVL NORTH PC	06-19-2002	Wed	10:10	0.46	50	117.00	350.00	68.00	42.44	85.75
52	73-02	IKON 2ND LVL NORTH PC	06-20-2002	Thu	09:44	0.71	50	131.00	264.00	92.00	27.52	58.48
53	74-02	IKON 2ND LVL EAST PC	06-24-2002	Mon	11:24	0.46	71	118.00	330.00	68.00	40.02	80.85
54	77-02	IKON 2ND LVL NORTH PC	07-01-2002	Mon	11:58	1.27	50	121.00	279.00	71.00	33.11	67.38
55	78-02	IKON 1ST LVL WEST PC	07-02-2002	Tue	10:02	0.37	33	116.00	283.00	30.00	51.67	91.08
56	79-02	IKON 2ND LVL EAST PC	07-03-2002	Wed	11:18	0.28	45	114.00	420.00	73.00	49.16	100.49
57	80-02	IKON 1ST LVL WEST PC	07-08-2002	Mon	02:50	0.44	63	119.00	243.00	33.00	42.30	75.76
58	81-02	IKON 1ST LVL WEST PC	07-09-2002	Tue	10:30	0.59	71	116.00	268.00	33.00	46.65	83.55
59	82-02	IKON 2ND LVL NORTH PC	07-09-2002	Tue	12:58	0.47	63	118.00	278.00	36.00	46.33	84.19
60	83-02	IKON 1ST LVL WEST PC	07-11-2002	Thu	10:34	0.58	56	118.00	231.00	33.00	40.21	72.02
61	84-02	IKON 1ST LVL WEST PC	07-15-2002	Mon	03:26	0.48	31	116.00	217.00	33.00	37.77	67.65
62	85	IKON 2ND LVL FLOOR	07-17-2002	Wed	08:59	0.28	45	116.00	420.00	20.00	93.91	154.73
63	86	IKON 2ND LVL NORTH PC	07-17-2002	Wed	10:59	0.46	71	114.00	255.00	33.00	44.39	79.50
64	87-02	IKON 1ST LVL WEST PC	07-18-2002	Thu	11:38	0.65	56	116.00	223.00	33.00	38.82	69.52
65	89-02	IKON TOP LVL N QY PC	07-22-2002	Mon	12:03	0.11	56	112.00	777.00	37.00	127.74	233.18
66	90-02	IKON 2ND LVL WEST PC	07-23-2002	Tue	10:47	0.85	56	119.00	228.00	36.00	38.00	69.05
67	91-02	IKON BTM LVL WEST PC	07-24-2002	Wed	10:17	0.07	56	106.00	420.00	35.00	70.99	128.40
68	92-02	IKON 2ND LVL WEST PC	07-26-2002	Fri	11:27	0.90	71	120.00	215.00	32.00	38.01	67.72
69	94-02	IKON BTM LVL WEST PC	07-30-2002	Tue	12:45	0.17	38	114.00	431.00	36.00	71.83	130.53
70	95-02	IKON 2ND LVL WEST PC	08-02-2002	Fri	10:49	0.80	71	121.00	158.00	58.00	20.75	40.82
71	97-02	IKON BTM LVL WEST PC	08-07-2002	Wed	10:56	0.28	50	114.00	392.00	36.00	65.33	118.72
72	98-02	IKON BOTTOM LEVEL	08-08-2002	Thu	11:02	0.14	56	114.00	414.00	36.00	69.00	125.38
73	99-02	IKON BTM LVL WEST PC	08-09-2002	Fri	09:32	0.11	100	114.00	460.00	36.00	76.67	139.31
74	100-02	IKON 2ND LEVEL	08-12-2002	Mon	10:42	1.04	56	122.00	160.00	70.00	19.12	38.82
75	101-02	IKON BTM LVL WEST PC	08-14-2002	Wed	09:31	0.10	26	114.00	450.00	35.00	76.06	137.57
76	103-02	IKON 3RD LEVEL	08-16-2002	Fri	10:01	0.11	50	109.00	370.00	38.00	60.02	110.05
77	103-02	IKON 3RD LEVEL	08-16-2002	Fri	10:01	0.14	26	112.00	320.00	38.00	51.91	95.18
78	105-02	IKON 2ND LVL WEST PC	08-19-2002	Mon	11:30	1.16	45	123.00	160.00	72.00	18.86	38.46
79	106-02	IKON BTM LVL WEST PC	08-20-2002	Tue	10:50	0.14	38	114.00	450.00	41.00	70.28	130.50
80	107-02	IKON BTM LVL SOUTH PC	08-21-2002	Wed	11:27	0.19	50	106.00	376.00	43.00	57.34	107.32
81	109-02	IKON BTM LVL WEST PC	08-23-2002	Fri	11:44	0.35	100	109.00	475.00	86.00	51.22	107.61
82	111-02	IKON BTM LVL SOUTH PC	08-27-2002	Tue	10:30	0.22	167	109.00	375.00	43.00	57.19	107.04
83	112-02	IKON BTM LVL WEST PC	08-28-2002	Wed	10:02	0.24	45	116.00	120.00	43.00	18.30	34.25
84	113-02	IKON 2ND LVL WEST PC	08-30-2002	Fri	10:29	1.19	71	124.00	185.00	28.00	34.96	60.92
85	114-02	IKON BTM LVL SOUTH PC	09-03-2002	Tue	10:45	0.04	1	100.00	360.00	43.00	54.90	102.76
86	121-02	IKON BTM LVL WEST PC	09-13-2002	Fri	12:13	0.49	45	114.00	390.00	43.00	59.47	111.32

Precision Blasting  
 Services, Inc.

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V5.00 Page: 3  
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 Coefficient of determination ( $r^2$  - "goodness of fit") = 0.435 Standard deviation = 0.247

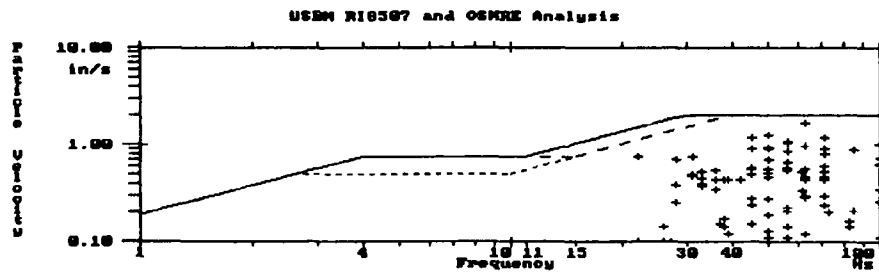
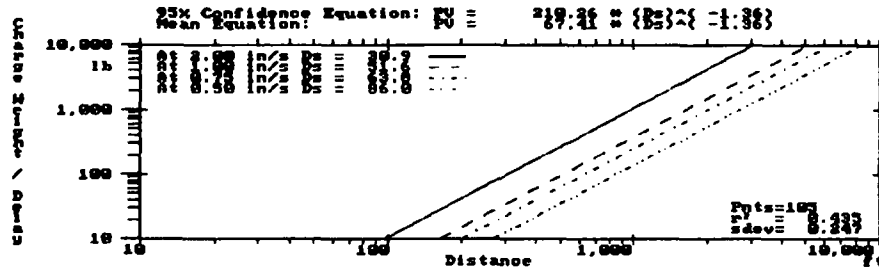
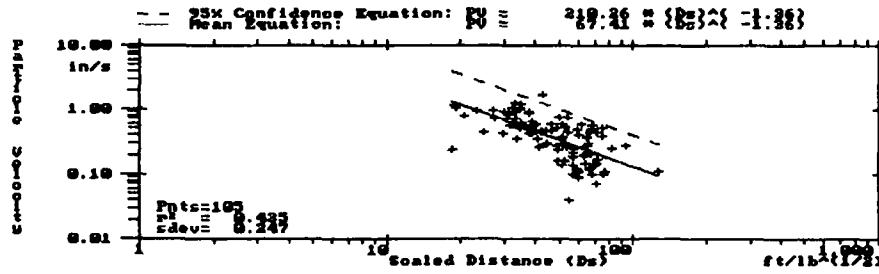
No.	SHOT	LOCATION	DATE	DAY	TIME	P.P.V.	FREQUENCY	AIR	DISTANCE	CHARGE	SCALED DISTANCES		
						in/s	Hz	PRESSURE	ft	WEIGHT	VIBRAT.	AIRPRES.	
								dB		lb			
87	123-02	IKOH	BTM LVL SOUTH PC	09-17-2002	Tue	11:48	0.29	63	114.00	343.00	43.00	52.31	97.90
88	124-02	IKOH	BTM LVL SOUTH PC	09-19-2002	Thu	11:50	0.21	56	112.00	431.00	44.00	64.98	122.08
89	126-02	IKOH	BTM LVL WEST PC	09-23-2002	Mon	12:40	0.12	39	114.00	392.00	43.00	59.78	111.89
90	127-02	IKOH	BOTTOM LEVEL	09-24-2002	Tue	12:07	0.20	73	109.00	422.00	44.00	63.62	119.53
91	128-02	IKOH	2ND LVL NORTH PC	09-25-2002	Wed	10:25	0.89	85	125.00	202.00	37.00	33.21	60.62
92	131-02	IKOH	BTM LVL SOUTH PC	09-30-2002	Mon	10:13	0.30	63	109.00	435.00	44.00	65.58	123.22
93	132-02	IKOH	BTM LVL WEST PC	10-02-2002	Wed	10:05	0.98	0	122.00	210.00	37.00	34.52	63.02
94	137-02	IKOH	BTM LVL SOUTH PC	10-09-2002	Wed	11:44	0.24	71	112.00	331.00	44.00	49.90	93.76
95	139-02	IKOH	BTM LVL WEST PC	10-14-2002	Mon	10:55	0.81	71	122.00	190.00	37.00	31.24	57.02
96	140-02	IKOH	BOTTOM LVL SOUTH	10-15-2002	Tue	11:44	0.21	85	84.00	365.00	29.00	67.78	118.80
97	141-02	IKOH	BTM LVL WEST PC	10-17-2002	Thu	12:05	0.57	63	118.00	190.00	34.00	32.58	58.65
98	142-02	IKOH	BTM LVL SOUTH PC	10-18-2002	Fri	11:46	0.26	125	112.00	325.00	37.00	53.43	97.53
99	145-02	IKOH	BTM LVL WEST PC	10-23-2002	Wed	11:26	1.06	56	119.00	211.00	37.00	34.69	63.32
100	146-02	IKOH	BTM LVL WEST PC	10-25-2002	Fri	10:16	0.47	50	112.00	271.00	49.00	38.71	74.06
101	148-02	IKOH	BTM LVL WEST PC	10-29-2002	Tue	10:17	0.42	125	114.00	241.00	41.00	37.64	69.89
102	149-02	IKOH	BTM LVL WEST PC	10-30-2002	Wed	10:11	0.60	71	121.00	326.00	40.00	51.55	95.32
103	150-02	IKOH	BTM LVL WEST PC	10-31-2002	Thu	10:25	0.62	100	114.00	232.00	44.00	34.98	65.72
104	29-02	IKOH	BTM LVL WEST PC	01-09-2003	Thu	11:31	0.38	28	114.00	475.00	40.00	75.10	138.89
105	75-02	IKOH	2ND LVL NORTH PC	01-13-2003	Mon	12:39	1.68	63	119.00	365.00	72.00	43.02	87.74

Precision Blasting  
 Services, Inc.

VIBRATION CONSULTANT

V5.00 Page: 4  
 Date: 05-20-2003

Number of records: 105 File: PPTILIKB.VIB 2003-Jan-15 Wed 11:02 9,640  
 INDIAN POINT



APPENDIX 2 – BLAST DESIGNER PROGRAM

Precision Blasting  
Services, Inc.

B L A S T  
D E S I G N E R

V5.01 Page: 1  
Date: 05-20-2003

INDIAN POINT

INPUT DATA OF ENTRY 1 Column loaded borehole with one explosive

Fixed production ..... 100 holes  
Initiation ..... Delayed  
Total number of holes ..... 100  
Diameter of blast hole ..... 3.00 in  
Hole depth ..... 26.78 ft  
Bench height ..... 25.00 ft  
Subdrill ..... 1.78 ft  
Burden ..... 5.94 ft  
Spacing ..... 8.31 ft  
Stemming (chips) ..... 4.16 ft  
Number of primers ..... 1 / hole  
Number of initiators ..... 1 / hole  
Number of surface delays ..... 1 / hole  
Delay time along the row ..... 25 ms  
Delay time between rows ..... 100 ms  
Column explosive type ..... EMULSION  
Expl. charge diameter ..... 2.50 in  
Expl. spec. gravity ..... 1.20  
Expl. bulk strength ..... 0.00  
Loading density ..... 2.55 lb/ft  
Weight of explosive ..... 57.78 lb  
Length of explosive ..... 22.63 ft  
Rock type ..... DIABASE  
Rock spec. gravity ..... 2.80 g/cm3

Precision Blasting  
 Services, Inc.

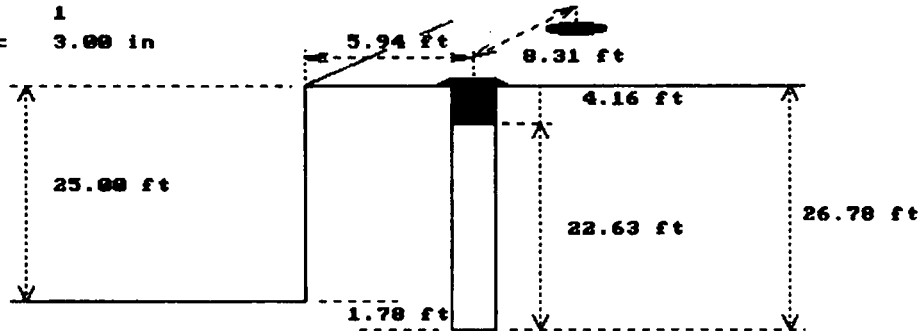
B L A S T  
 D E S I G N E R

V5.01 Page: 2  
 Date: 05-20-2003

INDIAN POINT

RESULTS OF ENTRY 1 Column loaded borehole with one explosive

ENTRY 1  
 Diameter = 3.00 in



Total drill length	.....	2,678.12 ft
Powder factor	.....	1.26 lb/yd <sup>3</sup>
Powder ratio	.....	0.79 yd <sup>3</sup> /lb
Total explosive	.....	5,777.74 lb
Tot. volume produced	.....	4,569.70 yd <sup>3</sup>
Tot. weight produced	.....	10,783.48 ton



Precision Blasting Services, Inc. B L A S T  
D E S I G N E R V5.01 Page: 3  
Date: 05-20-2003

E N T R Y	1	2	3	4	units
-----					
File name:					
-----					
Initiation	Delayed				
Top explosive	EMULSION				
Expl. charge diamete	2.50				in
Expl. spec. gravity	1.20				g/cm3
Expl. bulk strength					
Powder column length	22.63				ft
Loading density	2.55				lb/ft
Explosive weight	57.78				lb
Top rock type	DIABASE				
Rock spec. gravity	2.80				g/cm3
-----					
Bottom explosive					
Expl. charge diamete					in
Expl. spec. gravity					g/cm3
Expl. bulk strength					
Explosive weight					lb
Loading density					lb/ft
Explosive weight					lb
Bottom rock type					
Rock spec. gravity					g/cm3
-----					
Number of holes	100.00				
Hole diameter	3.00				in
Bench height	25.00				ft
Hole depth	26.78				ft
Subdrill	1.78				ft
Drilling angle					
Burden	5.94				ft
Spacing	8.31				ft
Stemming (chips)	4.16				ft
Stemming (dust)					
Total drill length	2,678.12				ft
Powder factor	1.26				lb/yd3
Powder ratio	0.79				yd3/lb
Total col. explosive	5,777.74				lb
Total base explosive					
Total explosive	5,777.74				lb
Tot. volume produced	4,569.70				yd3
Tot. weight produced	10,783.48				ton

-----END OF SUMMARY REPORT-----

APPENDIX 3 – BLAST COST ANALYST PROGRAM

Precision Blasting  
 Services, Inc.

BLASTING COST ANALYST

V5.01 Page: 1  
 Date: 05-20-2003

INDIAN POINT

ENTRY 1 Column loaded borehole

P A T T E R N

Total number of holes ..... 100  
 Diameter of the blasthole ..... 3.00 in  
 Burden ..... 6.00 ft  
 Spacing ..... 8.00 ft  
 Stemming ..... 8.00 ft  
 Subdrill ..... 2.00 ft  
 Bench height ..... 25.00 ft  
 Hole depth ..... 27.00 ft

Rock type ..... DIABASE  
 Specific gravity of the rock ..... 2.80 g/cm3

D R I L L I N G

Total drill length ..... 2,700.00 ft  
 Drilling cost per length ..... 2.00 \$/ft  
 Total drill cost ..... 5,400.00 \$  
 Drill factor ..... 0.61 ft/yd3  
 Drilling cost per volume ..... 1.22 \$/yd3  
 Drilling cost per weight ..... 0.51 \$/ton

E X P L O S I V E S

Type (brand) of explosive . . . . . EMULSION  
 Specific gravity of explosive..... 1.20 g/cm3  
 Diameter of explosive..... 2.50 in  
 Energy..... 0.00  
 Cost of explosive..... 1.5000 \$/lb

Total weight of explosive ..... 4,851.98 lb  
 Total cost of explosive / volume . . 1.64 \$/yd3  
 Total cost of explosive / weight . . 0.69 \$/ton  
 Powder factor . . . . . 1.09 lb/yd3  
 . . . . . 0.92 yd3/lb

Total cost of explosive . . . . . 7,277.97 \$  
 Energy . . . . . 0.00 /yd3  
 Energy . . . . . 0.00 /ton

Initiators	Delay ms	Quantity	Cost/Unit	Cost
1 DUAL DELAY	25,350	6	6.00	36.00
Total initiator cost:				36.00

Rock volume . . . . . 4,444.44 yd3  
 Rock weight . . . . . 10,487.90 ton

Precision Blasting  
 Services, Inc.

BLASTING COST ANALYST

V5.01 Page: 2  
 Date: 05-20-2003

INDIAN POINT

ENTRY 1 Column loaded borehole

Category	Total cost \$	Percent %	Cost/weight \$/ton	Cost/volume \$/yd3
Drill cost	5,400.00	41.49	0.515	1.215
Explosive cost	7,277.97	55.92	0.694	1.638
Initiator cost	36.00	0.28	0.003	0.008
Primer cost				
Surface delay cost				
Sleeves cost				
Delivery cost				
Protection cost				
Pumping cost				
Service cost	300.00	2.31	0.029	0.068
Seismic monitor cost				
Labor cost				
Other costs				
<b>Total cost</b>	<b>13,013.97</b>		<b>1.241</b>	<b>2.928</b>

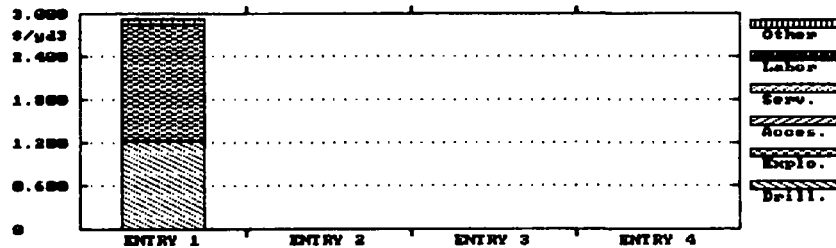
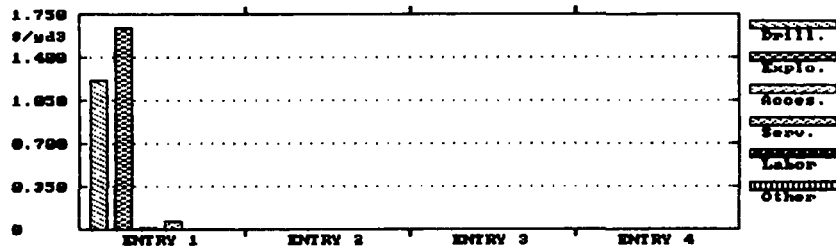
Precision Blasting Services, Inc.		BLASTING COST ANALYST			V5.01 Page: 3
					Date: 05-20-2003
ENTRY	1	2	3	4	unit
<b>File</b>					
Number of holes	100				
Hole diameter	3.00				in
Burden	6.00				ft
Spacing	8.00				ft
Stemming	8.00				ft
Subdrill	2.00				ft
Bench height	25.00				ft
Hole depth	27.00				ft
Rock type	DIABASE				
Spec. gr. rock	2.80				g/cm3
Drill length	2,700.00				ft
Drill cost/vol.	1.22				\$/yd3
Drill cost/wgh.	0.51				\$/ton
Drill factor	0.61				ft/yd3
Drill cost	5,400.00				\$
Explosive A:	EMULSION				
B:					
C:					
Expl. weight	4,851.98				lb
Expl. cost/vol.	1.64				\$/yd3
Expl. cost/wgh.	0.69				\$/ton
Powder factor	1.09				lb/yd3
	0.92				yd3/lb
Expl. cost	7,277.97				\$
Energy					/yd3 /ton
Initiator cost	36.00				\$
Primer cost					\$
Surface delay					\$
Sleeve cost					\$
Delivery cost					\$
Protection cost					\$
Pumping cost					\$
Shot service	300.00				\$
Seismic monitor					\$
Labor cost					\$
Other cost					\$
Rock volume	4,444.44				yd3
	10,487.90				ton
Cost	2.93				\$/yd3
	1.24				\$/ton
<b>Total cost</b>	<b>13,013.97</b>				<b>\$</b>

Precision Blasting  
 Services, Inc.

BLASTING COST ANALYST

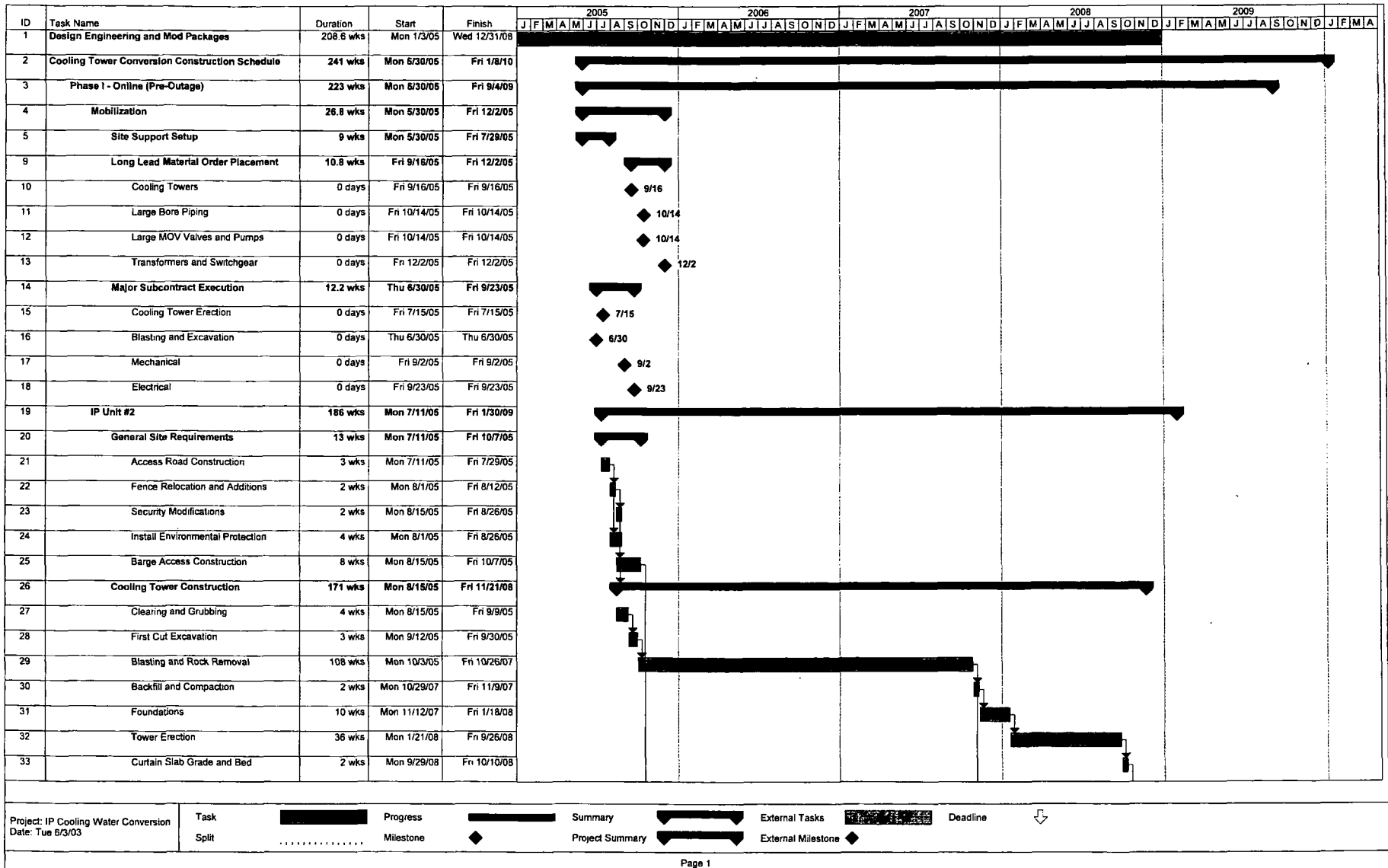
V5.01 Page: 4  
 Date: 05-20-2003

ENTRY	1	2	3	4	unit
File name:					
Drill cost	1.22				\$/yd3
Expl. cost	1.64				\$/yd3
Accessories	0.01				\$/yd3
Services	0.07				\$/yd3
Labor cost					\$/yd3
Other cost					\$/yd3
Total cost	2.93				\$/yd3



APPENDIX 6B – COOLING SYSTEM CONVERSION SCHEDULE

Economic and Environmental Impacts Associated with  
Conversion of Indian Point Units 2 and 3 To A Closed-Loop  
Condenser Cooling Water Configuration











Economic and Environmental Impacts Associated with  
Conversion of Indian Point Units 2 and 3 To A Closed-Loop  
Condenser Cooling Water Configuration

---

Attachment 7  
Electrical Distribution Model Output Reports

### ETAP® PowerStation ® Model Results

This attachment is provided to model the anticipated electrical distribution system required to support conversion of Indian Point Units 2 and 3 to a closed loop condenser cooling water configuration, and account for expected electrical parasitic losses due to the new components. The following documents are included in this attachment (for Indian Point 2 and Indian Point 3):

One Line Sketches of the proposed Distribution System as modeled including,

- 138kV One Line Distribution for Indian Point 2 and Indian Point 3
- One Line Distribution for Indian Point 2
- One Line Distribution for Indian Point 3

Analytical evaluation via ETAP PowerStation model of projected plant power demand using load flow and voltage drop calculations. The evaluation is presented as excerpts from the output reports of two loading configurations:

- Load Flow and voltage drop of tower fan and circulating water pump loads at full load.
- Load Flow and voltage drop of tower fan and circulating water pump loads at the reduced load representative of wet cycle fans and circulating water pumps only.

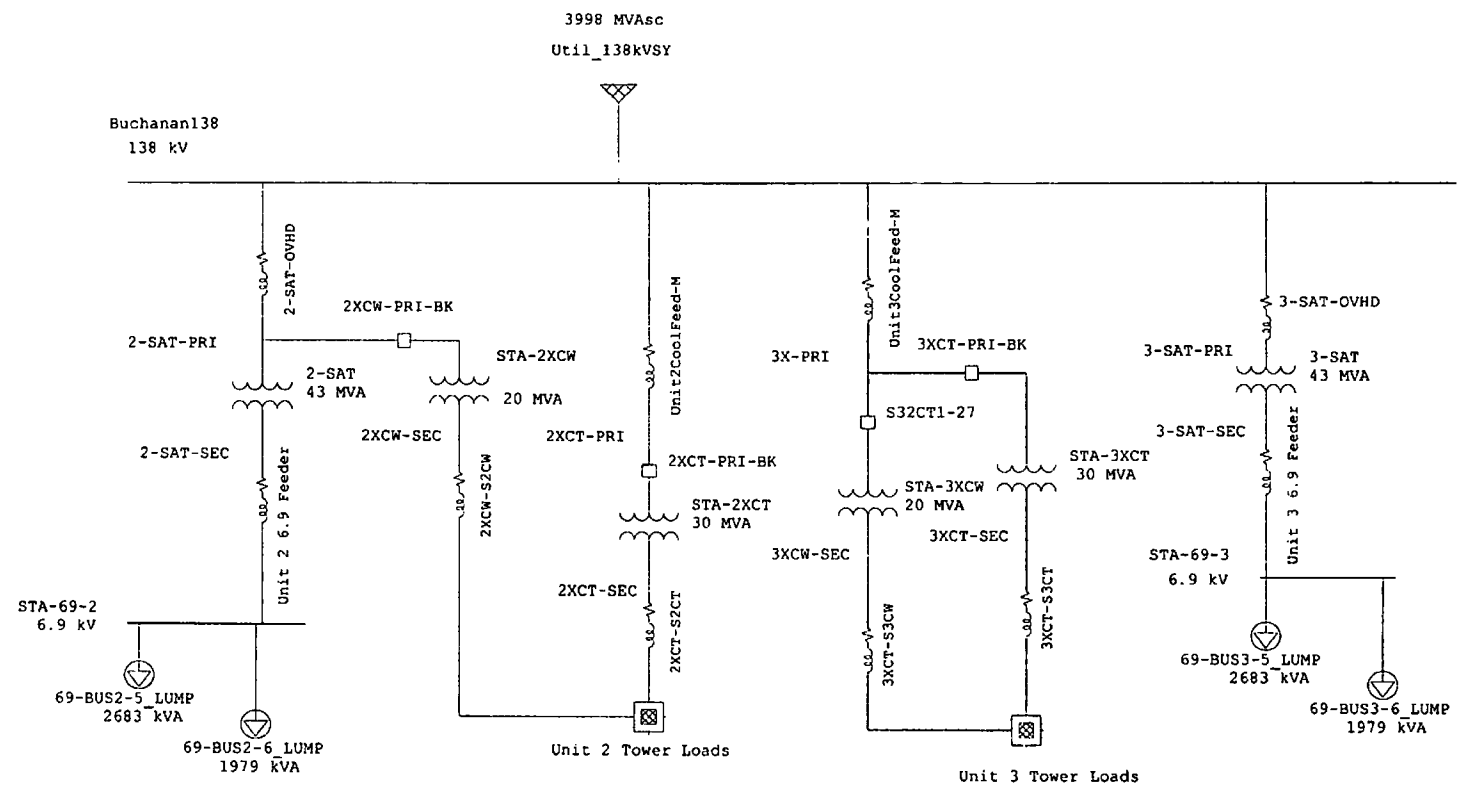
Analytical evaluation via ETAP PowerStation model of projected plant short circuit burden using IEEE methodology and calculations. The evaluation is presented as excerpts from the output report which assumed maximum fault and rated loading conditions.

The software model was developed and run using the following assumed parameters:

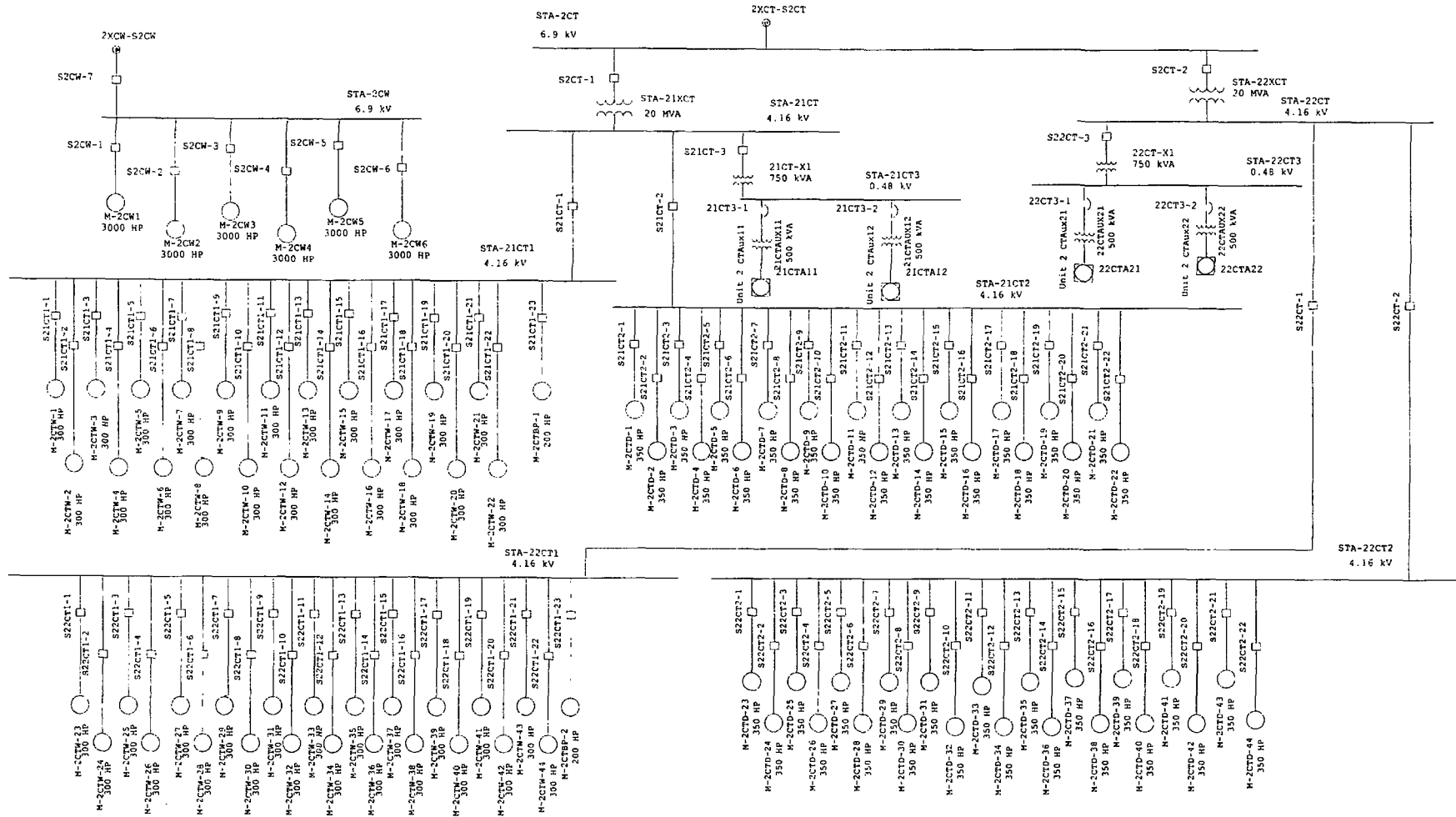
- Motors, cables, and transformer characteristics were sized based upon preliminary vendor information of tower configuration and required horsepower. Subsequent analytical parameters were assumed based upon the standard or typical values available in the software database for the input size of each component.
- The 138kV Buchanan Substation parameters were taken from a representative Indian Point 2 calculation (FEX-00143-00). Likewise, Indian Point 2 loads on the Buchanan Substation during normal operation were derived from the same calculation by assuming lumped load groups equivalent to the Indian Point 2 calculated load. Based upon plant similarities, the same assumed load was used for Indian Point 3.
- The fans for dry cycle cooling were either assumed to be on at full load (350HP) during nominal conditions, or off during wet cycle only conditions. Reduction in parasitic load due to variable speed dry cycle fan motors is not considered directly in this analysis, rather, it is accounted for in the percent of the time the tower is assumed to be in either full load or wet cycle only conditions.

Power to circulating water loads passes through transformers STA-2XCW and STA-3XCW. Power to tower loads (wet and dry fans and booster pumps) passes through transformers STA-2XCT and STA-3XCT. The expected power flow is highlighted in the report excerpts.

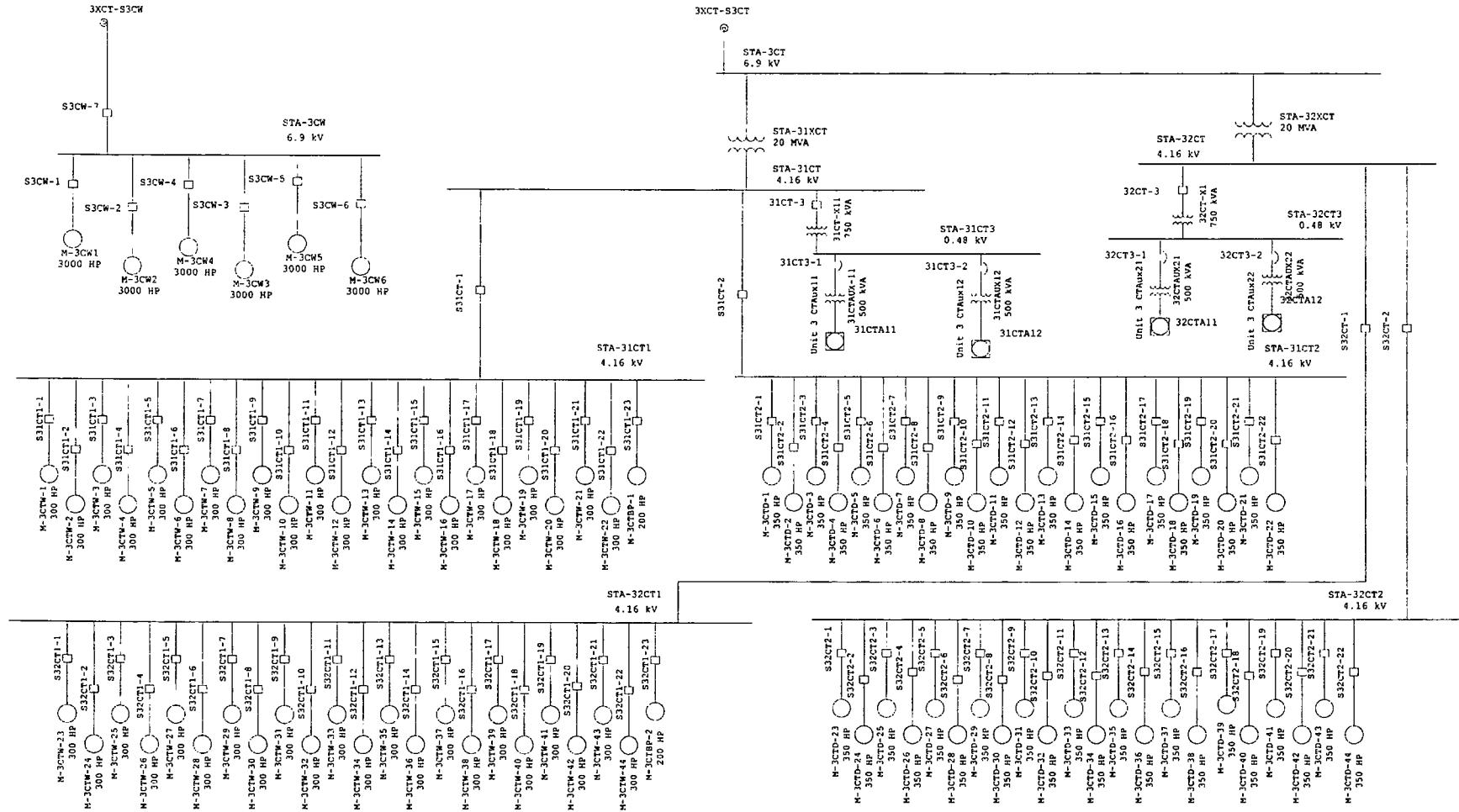
One-Line Diagram - IP-138kV



One-Line Diagram - IP-138kV=>Unit 2 Tower Loads



One-Line Diagram - IP-138kV=>Unit 3 Tower Loads





Economic and Environmental Impacts Associated with  
 Conversion of Indian Point Units 2 and 3 To A Closed-Loop  
 Condenser Cooling Water Configuration

SYSTEM ANALYSIS

Project: IP Load Flow  
 Location: IP 2 & 3  
 Contract: CO199  
 Engineer: Eric J Praser

=====  
 PowerStation 4.0.4C  
 Study Case: LF -FullLoad

Page: 1  
 Date: 06-02-2003  
 SN: ENERCONSV  
 File: IndianPoint

-----  
 Indian Point 2 & 3 Cooling Tower  
 Wet and Dry loads at nominal.  
 =====

Electrical Transient Analyzer Program  
 -----

LOAD FLOW ANALYSIS  
 Loading Category 2 ( Normal )

Normal Loading  
 -----

	Swing	Gen.	Load	Total
	----	----	----	----
Number of Buses:	1	0	40	41

	XFRM2	React.	Line/Cable	Imp.	Tie PD	XFRM3	Total
	----	----	----	----	----	----	----
Number of Branches:	22	0	10	0	8	0	40

Maximum Number of Iterations: 99  
 Precision of the Solution: .00010 MW and Mvar  
 Method of Solution: Newton-Raphson  
 System Frequency: 60.0 Hz  
 Unit System: English  
 Data Filename: IndianPoint  
 Output Filename: L:\PROJECTS\CO\CO199\BYRON-ELECTRICAL\IPMODEL\FullLoad.lfr

Economic and Environmental Impacts Associated with  
 Conversion of Indian Point Units 2 and 3 To A Closed-Loop  
 Condenser Cooling Water Configuration

Attachment 7  
 Section 2 – Load Flow (Excerpts)

LOAD FLOW REPORT

Project: IP Load Flow  
 Location: IP 2 & 3  
 Contract: CO199  
 Engineer: Eric J Praser

PowerStation 4.0.4C  
 Study Case: LF -FullLoad

Page: 11  
 Date: 06-02-2003  
 SN: ENERCONSV  
 File: IndianPoint

Indian Point 2 & 3 Cooling Tower  
 Wet and Dry loads at nominal.

Bus Information & Nom kV			Voltage		Generation		Motor Load		Static Load		Load Flow					XFRM
ID	Type	kV	% Mag.	Ang.	MW	Mvar	MW	Mvar	MW	Mvar	To Bus ID	MW	Mvar	Amp	%PF	% Tap
2-SAT-PRI	Load	138.00	99.97	0.0	0.00	0.00	0.00	0.00	0.00	0.00	Buchanan138	-17.56	-10.92	86	84.9	
											2-SAT-SEC	4.09	2.56	20	84.8	-2.500
											2XCW-SEC	13.47	8.35	66	85.0	-2.500
2-SAT-SEC	Load	6.90	101.58	-0.8	0.00	0.00	0.00	0.00	0.00	0.00	STA-69-2	4.09	2.48	393	85.5	
											2-SAT-PRI	-4.09	-2.48	393	85.5	
2XCT-PRI	Load	138.00	99.96	0.0	0.00	0.00	0.00	0.00	0.00	0.00	Buchanan138	-23.99	-13.57	115	87.0	
											2XCT-SEC	23.99	13.57	115	87.0	-2.500
2XCT-SEC	Load	6.90	100.96	-4.0	0.00	0.00	0.00	0.00	0.00	0.00	STA-2CT	23.90	11.40	2194	90.3	
											2XCT-PRI	-23.90	-11.40	2194	90.3	2.500
2XCW-SEC	Load	6.90	101.19	-3.3	0.00	0.00	0.00	0.00	0.00	0.00	STA-2CW	13.41	7.28	1261	87.9	
											2-SAT-PRI	-13.41	-7.28	1261	87.9	2.500
3-SAT-PRI	Load	138.00	99.99	0.0	0.00	0.00	0.00	0.00	0.00	0.00	Buchanan138	-4.09	-2.56	20	84.8	
											3-SAT-SEC	4.09	2.56	20	84.8	-2.500
3-SAT-SEC	Load	6.90	101.61	-0.8	0.00	0.00	0.00	0.00	0.00	0.00	STA-69-3	4.09	2.48	393	85.5	
											3-SAT-PRI	-4.09	-2.48	393	85.5	
3XCT-SEC	Load	6.90	100.97	-4.0	0.00	0.00	0.00	0.00	0.00	0.00	STA-3CT	23.92	11.48	2199	90.2	
											3X-PRI	-23.92	-11.48	2199	90.2	2.500
3XCW-SEC	Load	6.90	101.13	-3.3	0.00	0.00	0.00	0.00	0.00	0.00	STA-3CW	13.42	7.31	1263	87.8	
											3X-PRI	-13.42	-7.31	1263	87.8	2.500
3X-PRI	Load	138.00	99.93	0.0	0.00	0.00	0.00	0.00	0.00	0.00	Buchanan138	-37.47	-22.04	181	86.2	
											3XCT-SEC	24.00	13.65	115	86.9	-2.500
											3XCW-SEC	13.47	8.38	66	84.9	-2.500
21CTA11	Load	0.12	96.53	-10.1	0.00	0.00	0.00	0.00	0.25	0.12	STA-21CT3	-0.25	-0.12	1396	90.0	
21CTA12	Load	0.12	96.06	-10.1	0.00	0.00	0.18	0.11	0.07	0.04	STA-21CT3	-0.25	-0.15	1472	85.0	
22CTA21	Load	0.12	101.62	-6.0	0.00	0.00	0.00	0.00	0.00	0.00	STA-22CT3	0.00	0.00	2	0.0	
22CTA22	Load	0.12	101.62	-6.0	0.00	0.00	0.00	0.00	0.00	0.00	STA-22CT3	0.00	0.00	4	0.0	
31CTA11	Load	0.12	96.82	-10.2	0.00	0.00	0.00	0.00	0.25	0.12	STA-31CT3	-0.25	-0.12	1400	90.0	

\* A regulated (constant voltage) bus.

LOAD FLOW SUMMARY

Project: IP Load Flow  
 Location: IP 2 & 3  
 Contract: C0199  
 Engineer: Eric J Praser

=====  
 PowerStation 4.0.4C  
 Study Case: LF -FullLoad

Page: 23  
 Date: 06-02-2003  
 SN: ENERCONSV  
 File: IndianPoint

-----  
 Indian Point 2 & 3 Cooling Tower  
 Wet and Dry loads at nominal.  
 =====

SUMMARY OF TOTAL GENERATION, LOADING & DEMAND  
 -----

	MW	Mvar	MVA	% PF
	=====	=====	=====	=====
Swing Bus(es):	83.150	49.118	96.574	86.1 Lagging
Generators:	0.000	0.000	0.000	100.0 Lagging
Total Demand:	83.150	49.118	96.574	86.1 Lagging
Total Motor Load:	80.213	38.626	89.029	90.1 Lagging
Total Static Load:	2.444	1.422		
Apparent Losses:	0.493	9.070		
System Mismatch:	0.000	0.000		

Number of Iterations = 3

Economic and Environmental Impacts Associated with  
 Conversion of Indian Point Units 2 and 3 To A Closed-Loop  
 Condenser Cooling Water Configuration

SYSTEM ANALYSIS

Project: IP Load Flow  
 Location: IP 2 & 3  
 Contract: CO199  
 Engineer: Eric J Praser

=====  
 PowerStation 4.0.4C  
 Study Case: LF-WetLoad

Page: 1  
 Date: 06-02-2003  
 SN: ENERCONSV  
 File: IndianPoint

-----  
 Indian Point 2 & 3 Cooling Tower  
 Dry Loads off, Wet cycle loads at nominal.  
 =====

Electrical Transient Analyzer Program  
 -----

LOAD FLOW ANALYSIS  
 Loading Category 8 ( Wet Only )

Normal Loading  
 -----

	Swing	Gen.	Load	Total			
	-----	-----	-----	-----			
Number of Buses:	1	0	40	41			

	XFRM2	React.	Line/Cable	Imp.	Tie PD	XFRM3	Total
	-----	-----	-----	-----	-----	-----	-----
Number of Branches:	22	0	10	0	8	0	40

Maximum Number of Iterations: 99  
 Precision of the Solution: .00010 MW and Mvar  
 Method of Solution: Newton-Raphson  
 System Frequency: 60.0 Hz  
 Unit System: English  
 Data Filename: IndianPoint  
 Output Filename: L:\PROJECTS\CO\CO199\BYRON-ELECTRICAL\IPMODEL\WetLoad.lfr

Economic and Environmental Impacts Associated with  
 Conversion of Indian Point Units 2 and 3 To A Closed-Loop  
 Condenser Cooling Water Configuration

Attachment 7  
 Section 2 – Load Flow (Excerpts)

LOAD FLOW REPORT

Project: IP Load Flow  
 Location: IP 2 & 3  
 Contract: C0199  
 Engineer: Eric J Praser

PowerStation 4.0.4C  
 Study Case: LF-WetLoad

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 File: IndianPoint

Indian Point 2 & 3 Cooling Tower  
 Dry Loads off, Wet cycle loads at nominal.

Bus Information & Nom kV			Voltage		Generation		Motor Load		Static Load		Load Flow					XFRM
ID	Type	kV	% Mag.	Ang.	MW	Mvar	MW	Mvar	MW	Mvar	To Bus ID	MW	Mvar	Amp	%PF	% Tap
2-SAT-PRI	Load	138.00	99.97	0.0	0.00	0.00	0.00	0.00	0.00	0.00	Buchanan138	-17.56	-10.92	86	84.9	
											2-SAT-SEC	4.09	2.56	20	84.8	-2.500
											2XCW-SEC	13.47	8.35	66	85.0	-2.500
2-SAT-SEC	Load	6.90	101.58	-0.8	0.00	0.00	0.00	0.00	0.00	0.00	STA-69-2	4.09	2.48	393	85.5	
											2-SAT-PRI	-4.09	-2.48	393	85.5	
2XCT-PRI	Load	138.00	99.98	0.0	0.00	0.00	0.00	0.00	0.00	0.00	Buchanan138	-10.67	-5.16	49	90.0	
											2XCT-SEC	10.67	5.16	49	90.0	-2.500
2XCT-SEC	Load	6.90	103.47	-1.7	0.00	0.00	0.00	0.00	0.00	0.00	STA-2CT	10.65	4.76	943	91.3	
											2XCT-PRI	-10.65	-4.76	943	91.3	2.500
2XCW-SEC	Load	6.90	101.19	-3.3	0.00	0.00	0.00	0.00	0.00	0.00	STA-2CW	13.41	7.28	1261	87.9	
											2-SAT-PRI	-13.41	-7.28	1261	87.9	2.500
3-SAT-PRI	Load	138.00	99.99	0.0	0.00	0.00	0.00	0.00	0.00	0.00	Buchanan138	-4.09	-2.56	20	84.8	
											3-SAT-SEC	4.09	2.56	20	84.8	-2.500
3-SAT-SEC	Load	6.90	101.61	-0.8	0.00	0.00	0.00	0.00	0.00	0.00	STA-69-3	4.09	2.48	393	85.5	
											3-SAT-PRI	-4.09	-2.48	393	85.5	
3XCT-SEC	Load	6.90	103.51	-1.7	0.00	0.00	0.00	0.00	0.00	0.00	STA-3CT	10.42	4.66	922	91.3	
											3X-PRI	-10.42	-4.66	922	91.3	2.500
3XCW-SEC	Load	6.90	101.16	-3.3	0.00	0.00	0.00	0.00	0.00	0.00	STA-3CW	13.42	7.31	1263	87.8	
											3X-PRI	-13.42	-7.31	1263	87.8	2.500
3X-PRI	Load	138.00	99.96	0.0	0.00	0.00	0.00	0.00	0.00	0.00	Buchanan138	-23.90	-13.43	114	87.2	
											3XCT-SEC	10.43	5.05	48	90.0	-2.500
											3XCW-SEC	13.47	8.38	66	84.9	-2.500
21CTA11	Load	0.12	105.30	-2.6	0.00	0.00	0.00	0.00	0.00	0.00	STA-21CT3	0.00	0.00	0	0.0	
21CTA12	Load	0.12	105.30	-2.6	0.00	0.00	0.00	0.00	0.00	0.00	STA-21CT3	0.00	0.00	0	0.0	
22CTA21	Load	0.12	105.30	-2.6	0.00	0.00	0.00	0.00	0.00	0.00	STA-22CT3	0.00	0.00	0	0.0	
22CTA22	Load	0.12	105.30	-2.6	0.00	0.00	0.00	0.00	0.00	0.00	STA-22CT3	0.00	0.00	0	0.0	
31CTA11	Load	0.12	105.81	-2.6	0.00	0.00	0.00	0.00	0.00	0.00	STA-31CT3	0.00	0.00	0	0.0	

\* A regulated (constant voltage) bus.

Economic and Environmental Impacts Associated with  
 Conversion of Indian Point Units 2 and 3 To A Closed-Loop  
 Condenser Cooling Water Configuration

Attachment 7  
 Section 2 – Load Flow (Excerpts)

Project: IP Load Flow  
 Location: IP 2 & 3  
 Contract: C0199  
 Engineer: Eric J Praser

=====  
 PowerStation 4.0.4C  
 Study Case: LF-WetLoad

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-----  
 Indian Point 2 & 3 Cooling Tower  
 Dry Loads off, Wet cycle loads at nominal.  
 -----

SUMMARY OF TOTAL GENERATION, LOADING & DEMAND  
 -----

	MW	Mvar	MVA	% PF
	=====	=====	=====	=====
Swing Bus(es):	56.236	32.086	64.746	86.9 Lagging
Generators:	0.000	0.000	0.000	100.0 Lagging
Total Demand:	56.236	32.086	64.746	86.9 Lagging
Total Motor Load:	54.234	27.444	60.782	89.2 Lagging
Total Static Load:	1.797	1.090		
Apparent Losses:	0.204	3.552		
System Mismatch:	0.000	0.000		

Number of Iterations = 3

SYSTEM ANALYSIS

Project: IP Load Flow  
 Location: IP 2 & 3  
 Contract: CO199  
 Engineer: Eric J Praser

PowerStation 4.0.4C  
 Study Case: SC

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-----  
 Indian Point 2 & 3 Cooling Tower  
 -----

-----  
 Electrical Transient Analyzer Program  
 -----

SHORT CIRCUIT ANALYSIS  
 -----

3-Phase, LG, LL, & LLG  
 1/2 Cycle (Momentary) Fault Currents

	Swing	Gen.	Load	Total
	-----	-----	-----	-----
Number of Buses:	1	0	40	41

	XFRM2	REACT.	LINE/CABLE	IMP.	TIE PD	XFRM3	TOTAL
	-----	-----	-----	-----	-----	-----	-----
Number of Branches:	22	0	10	0	8	0	40

	Synch. Gen.	Synch. Motor	Ind. Motor	Lump Motor	Uti- lity	Total
	-----	-----	-----	-----	-----	-----
Number of Machines:	0	0	288	6	1	295

System Frequency: 60.0 Hz

Unit System: English

Data File Name: IndianPoint

Output File Name: L:\PROJECTS\CO\CO199\BYRON-ELECTRICAL\IPMODEL\IP2&3.shr

**Economic and Environmental Impacts Associated with  
Conversion of Indian Point Units 2 and 3 To A Closed-Loop  
Condenser Cooling Water Configuration**

**Attachment 7  
Section 3 – Short Circuit (Excerpts)**

Project: IP Load Flow  
Location: IP 2 & 3  
Contract: CO199  
Engineer: Eric J Praser

S. C. SUMMARY REPORT  
PowerStation 4.0.4C  
Study Case: SC

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File: IndianPoint

Indian Point 2 & 3 Cooling Tower

1/2 Cycle - Three-Phase, LG, LL, & LLG Faults: ( Prefault Voltage = 100 % of the Bus Nominal Voltage)

Bus Information		3-Phase Fault			Line-to-Ground Fault			Line-to-Line Fault			Line-to-Line-to-Ground*		
ID	kV	Real	Imag.	Mag.	Real	Imag.	Mag.	Real	Imag.	Mag.	Real	Imag.	Mag.
2-SAT-PRI	138.00	1.344	-17.499	17.550	0.412	-4.736	4.754	15.155	1.164	15.199	-15.276	0.205	15.277
2-SAT-SEC	6.90	1.182	-23.835	23.864	1.111	-23.607	23.633	30.642	1.024	20.667	20.120	12.715	23.801
2XCT-PRI	138.00	1.348	-17.495	17.547	0.412	-4.736	4.754	15.151	1.167	15.196	-15.273	0.202	15.274
2XCT-SEC	6.90	1.833	-34.909	34.957	1.529	-31.582	31.619	30.232	1.588	30.274	29.583	16.004	33.635
2XCW-SEC	6.90	1.218	-26.425	26.453	1.120	-22.681	22.709	22.885	1.055	22.909	-23.400	8.878	25.027
3-SAT-PRI	138.00	1.371	-17.462	17.516	0.413	-4.735	4.753	15.122	1.187	15.169	-15.244	0.182	15.245
3-SAT-SEC	6.90	1.185	-23.837	23.867	1.113	-23.608	23.634	20.644	1.026	20.669	20.121	12.718	23.804
3XCT-SEC	6.90	1.668	-34.860	34.900	1.335	-31.560	31.588	30.189	1.444	30.224	29.645	15.859	33.621
3XCW-SEC	6.90	1.242	-26.308	26.337	1.131	-22.624	22.652	22.784	1.076	22.809	-23.301	8.846	24.923
3X-PRI	138.00	1.322	-17.532	17.582	0.410	-4.738	4.756	15.183	1.145	15.226	-15.304	0.225	15.306
21CTA11	0.12	2.303	-22.797	22.913	2.319	-25.911	26.014	19.743	1.994	19.843	18.627	16.993	25.214
21CTA12	0.12	3.993	-26.544	26.843	3.597	-29.054	29.276	22.988	3.458	23.247	21.513	19.473	29.017
22CTA21	0.12	2.189	-22.251	22.358	2.231	-25.436	25.534	19.270	1.896	19.363	18.179	16.734	24.708
22CTA22	0.12	2.189	-22.251	22.358	2.231	-25.436	25.534	19.270	1.896	19.363	18.179	16.734	24.708
31CTA11	0.12	2.302	-22.791	22.907	2.318	-25.906	26.009	19.738	1.994	19.838	18.622	16.991	25.209
31CTA12	0.12	3.992	-26.538	26.837	3.596	-29.049	29.271	22.983	3.457	23.241	21.508	19.470	29.012
32CTA11	0.12	2.188	-22.244	22.352	2.230	-25.431	25.528	19.264	1.895	19.357	18.174	16.731	24.702
32CTA12	0.12	2.188	-22.244	22.352	2.230	-25.431	25.528	19.264	1.895	19.357	18.174	16.731	24.702
Buchanan138	138.00	0.636	-18.444	18.455	0.364	-4.807	4.821	15.973	0.551	15.982	-16.086	0.831	16.107
STA-2CT	6.90	1.863	-34.755	34.805	1.920	-31.141	31.201	30.099	1.614	30.142	-31.060	12.489	33.477
STA-2CW	6.90	1.226	-26.365	26.394	1.318	-22.467	22.506	22.833	1.062	22.858	-23.496	8.723	25.063
STA-3CT	6.90	1.821	-34.118	34.166	3.108	-29.413	29.576	29.547	1.577	29.589	-31.425	11.300	33.395
STA-3CW	6.90	1.280	-26.017	26.049	2.044	-21.565	21.661	22.532	1.109	22.559	-23.702	8.075	25.040
STA-21CT	4.16	1.934	-32.581	32.638	2.044	-35.362	35.421	28.216	1.675	28.266	27.134	21.006	34.315
STA-21CT1	4.16	1.934	-32.581	32.638	2.044	-35.362	35.421	28.216	1.675	28.266	27.134	21.006	34.315
STA-21CT2	4.16	1.934	-32.581	32.638	2.044	-35.362	35.421	28.216	1.675	28.266	27.134	21.006	34.315
STA-21CT3	0.48	2.831	-15.649	15.903	2.776	-15.586	15.831	13.552	2.452	13.772	12.191	10.213	15.904
STA-22CT	4.16	1.902	-32.491	32.547	2.019	-35.292	35.349	28.138	1.647	28.186	27.064	20.957	34.229
STA-22CT1	4.16	1.902	-32.491	32.547	2.019	-35.292	35.349	28.138	1.647	28.186	27.064	20.957	34.229
STA-22CT2	4.16	1.902	-32.491	32.547	2.019	-35.292	35.349	28.138	1.647	28.186	27.064	20.957	34.229
STA-22CT3	0.48	2.471	-14.680	14.886	2.534	-14.931	15.145	12.713	2.140	12.892	-14.014	5.456	15.038
STA-31CT	4.16	1.905	-32.123	32.180	2.017	-34.909	34.968	27.820	1.650	27.868	26.750	20.762	33.862
STA-31CT1	4.16	1.905	-32.123	32.180	2.017	-34.909	34.968	27.820	1.650	27.868	26.750	20.762	33.862
STA-31CT2	4.16	1.905	-32.123	32.180	2.017	-34.909	34.968	27.820	1.650	27.868	26.750	20.762	33.862
STA-31CT3	0.48	2.828	-15.638	15.892	2.775	-15.579	15.824	13.543	2.449	13.763	12.182	10.209	15.894
STA-32CT	4.16	1.873	-32.034	32.089	1.992	-34.839	34.896	27.742	1.622	27.790	26.680	20.714	33.777
STA-32CT1	4.16	1.873	-32.034	32.089	1.992	-34.839	34.896	27.742	1.622	27.790	26.680	20.714	33.777
STA-32CT2	4.16	1.873	-32.034	32.089	1.992	-34.839	34.896	27.742	1.622	27.790	26.680	20.714	33.777

All fault currents are symmetrical momentary ( 1/2 cycle ) values in rms kA.  
\* LLG fault current is the larger of the two faulted line currents.



Economic and Environmental Impacts Associated with  
 Conversion of Indian Point Units 2 and 3 To A Closed-Loop  
 Condenser Cooling Water Configuration

Attachment 7  
 Section 3 – Short Circuit (Excerpts)

Project: IP Load Flow  
 Location: IP 2 & 3  
 Contract: CO199  
 Engineer: Eric J Praser

S. C. SUMMARY REPORT  
 =====  
 PowerStation 4.0.4C  
 Study Case: SC

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 File: IndianPoint

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 Indian Point 2 & 3 Cooling Tower  
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1/2 Cycle - Three-Phase, LG, LL, & LLG Faults: ( Prefault Voltage = 100 % of the Bus Nominal Voltage)

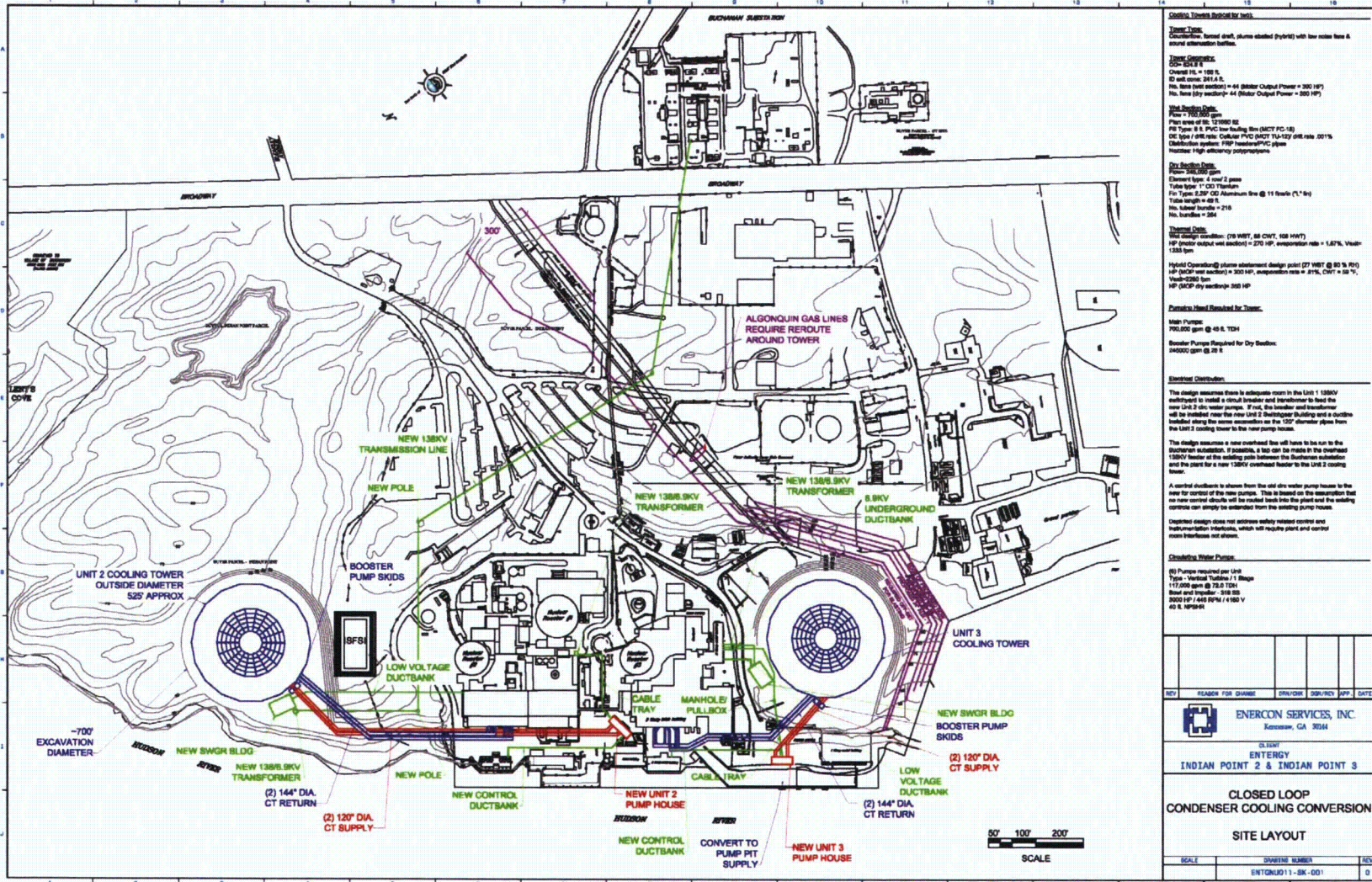
Bus Information		3-Phase Fault			Line-to-Ground Fault			Line-to-Line Fault			Line-to-Line-to-Ground*		
ID	kV	Real	Imag.	Mag.	Real	Imag.	Mag.	Real	Imag.	Mag.	Real	Imag.	Mag.
STA-32CT3	0.48	2.468	-14.669	14.875	2.532	-14.924	15.137	12.704	2.137	12.882	-14.004	5.456	15.029
STA-69-2	6.90	1.311	-23.115	23.152	2.144	-22.162	22.265	20.018	1.136	20.050	-21.440	9.476	23.441
STA-69-3	6.90	1.312	-23.578	23.614	1.266	-23.182	23.216	20.419	1.136	20.451	19.808	12.535	23.441

All fault currents are symmetrical momentary ( 1/2 cycle ) values in rms kA.  
 \* LLG fault current is the larger of the two faulted line currents.

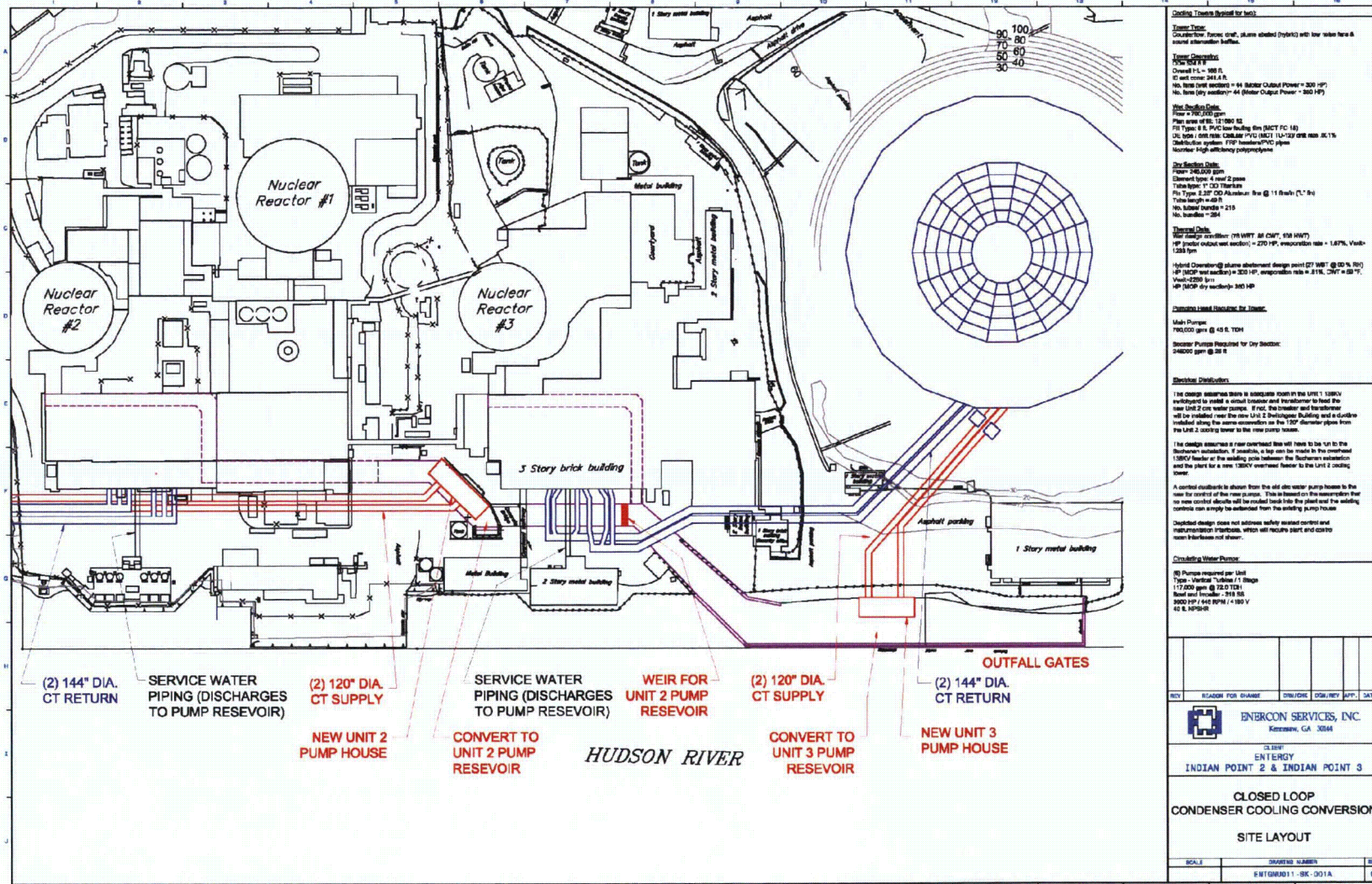


**Attachment 2**

**Post Modification Site Rendering and Conceptual Drawings**



<b>Coil Tower Support for Tank</b>			
<b>Tower Tank</b> Construction: Normal shell, alum. clad (py/pt) with low noise base & sound attenuation baffles.			
<b>Tower Capacity:</b> Cooling: 20,000 gpm Overhead HL = 100 ft. ID shell cone: 301.4 ft. No. fans (per section): 44 (Motor Output Power = 300 HP) No. fans (dry section): 44 (Motor Output Power = 300 HP)			
<b>Wind Section Data:</b> Flow: 110,000 gpm FE Type: 8" PVC fan building (MCT FC-10) DE type / drive ratio: Castles PVD DUCT 110-120 gpm rate 201% Distribution system: 150' fan/duct/pipe Nozzles: High efficiency polypropylene			
<b>Dry Section Data:</b> Flow: 200,000 gpm Element type: 4 and 2 gpm Tube spec: 1" OD Titanium FE Type: 2.0" OD Aluminum line @ 11 Rows (1" dia) Tube length = 49 ft. No. tubes/row = 216 No. bundles = 264			
<b>Thermal Data:</b> Wind design condition: (10 HWY, 68 CHY, 68 HWY) HP (motor output wet section) = 270 HP, evaporation rate = 1.67%, Vmax = 1250 gpm Hybrid Condition: (same as standard design point) (27 HWY @ 90 % RH) HP (MOP wet section) = 300 HP, evaporation rate = 31%, CHY = 58 °F, Vmax = 2300 gpm HP (MOP dry section) = 300 HP			
<b>Pumps to be Used Proposed for Tower:</b>			
Main Pumps: 700,000 gpm @ 45 ft. TDH			
Booster Pumps Required for Dry Section: 300,000 gpm @ 28 ft.			
<b>Electrical Distribution:</b>			
The design assumes there is adequate room in the Unit 1 138kV substation to install a circuit breaker and transformer to feed the new Unit 2-dry water pumps. If not, the breaker and transformer will be installed near the new Unit 2 Switchgear Building and a ductbank installed along the same excavation as the 120" diameter pipe from the Unit 3 cooling tower to the new pump house.			
The design assumes a new overhead line will have to be run to the Substation substation. If possible, a tap can be made in the overhead 138kV feeder at the existing pole between the Substation substation and the start for a new 138kV overhead feeder to the Unit 2 cooling tower.			
A control substation is shown from the old dry water pump house to the new for control of the new pumps. This is based on the assumption that no new control ducts will be needed back into the plant and the existing concrete can simply be extended from the existing pump house.			
Detailed design does not address safety related control and instrumentation requirements, which will require plant and control room interfaces not shown.			
<b>Condensing Water Pumps:</b>			
(4) Pumps required per Unit Type - Vertical Turbine 11 Stage 117,000 gpm @ 75.0 TDH Inlet and Impeller - 316 SS 3000 HP / 440 RPM / 1480 V 42 ft. height			
REV	READER FOR CHANGE	DATE	DATE
1			
 <b>ENERCON SERVICES, INC.</b> Kennesaw, GA 30144 CLIENT: <b>ENTERGY</b> <b>INDIAN POINT 2 &amp; INDIAN POINT 3</b> <b>CLOSED LOOP CONDENSER COOLING CONVERSION</b> <b>SITE LAYOUT</b>			
SCALE	DRAWING NUMBER	REV	
	ENTRINDU11-SK-001	0	



**Condition: Typical Typical for Unit 2**

**Design Data:**  
 Cooling Tower: Cross flow, slats shaded (typical) with low noise fans & sound attenuation baffles.  
 Fan: 100 HP  
 Overall P.L. = 100 ft.  
 Wet cross: 241.6 ft.  
 No. fans (dry weather) = 64 (Sector Output Power = 200 HP)  
 No. fans (dry weather) = 44 (Sector Output Power = 200 HP)

**Flow Schedule Data:**  
 Flow = 100,000 gpm  
 Pipe size of 18" (1200 ft)  
 Fit Type: 4.5" PVC low flange (NACE FC 14)  
 1/2" pipe + 4000 min. Coupling PVC (NACE) (1/2"-1/2" min. 10:1)  
 Distribution system: 1500 (NACE) PVC pipe  
 Material: High efficiency polypropylene

**Dry Section Data:**  
 Fan: 100,000 gpm  
 Element type: 4 row 2 pass  
 Tube type: 1" 20 Tube  
 Fit Type: 5.31" OD Aluminum: 1/2" @ 11 ft (1" @ 11 ft)  
 Tube length: 10 ft  
 No. tubes: 215  
 No. bundles: 204

**Thermal Data:**  
 Fan: 100,000 gpm  
 HP (motor output wet section) = 270 HP, evaporation rate = 1.87% (1.87% Wet-1.23% Dry)  
 HP (motor output wet section) = 270 HP, evaporation rate = 1.87% (1.87% Wet-1.23% Dry)

**Hybrid Operation (at alternate design point) (27 MBT @ 30% RW)  
 HP (motor output wet section) = 200 HP, evaporation rate = 2.1% (2.1% Wet-1.23% Dry)  
 HP (motor output wet section) = 200 HP, evaporation rate = 2.1% (2.1% Wet-1.23% Dry)**

**Condenser Load Schedule for Tower:**  
 Main Pump: 100,000 gpm @ 45 ft TDH  
 Sector Pump (Reserved for Dry Section): 240,000 gpm @ 30 ft

**Electrical Distribution:**  
 The design assumes there is adequate room in the Unit 2 138KV switchyard to install a second busbar and transformer to feed the new Unit 2 core water pump. If not, the transformer and busbar will be installed near the new Unit 2 Distribution Building and a ductbank installed along the same corridor as the 150" diameter pipe from the Unit 2 cooling tower to the new cooling tower.  
 The design assumes a new overhead line will have to be run to the Reactor station. If available, a line can be run to the overhead 138KV busbar of the existing area between the Reactor station and the plant for a new 138KV overhead feeder to the Unit 2 cooling tower.  
 A control building to house the old site water pump house to be used for control of the new pump. This is based on the assumption that no new control building will be added to the plant and the existing controls can simply be transferred from the existing pump house.  
 Detailed design does not address safety related control and instrumentation impacts, which will require plant and control room interfaces not shown.

**Condenser Water Pump:**  
 10 Pumps required per Unit  
 Type - Vertical Tubular / 1 Stage  
 17,000 gpm @ 22.5 TDH  
 Head and Impeller: 2018 SS  
 3000 HP / 640 RPM / 180 V  
 42 & 50000

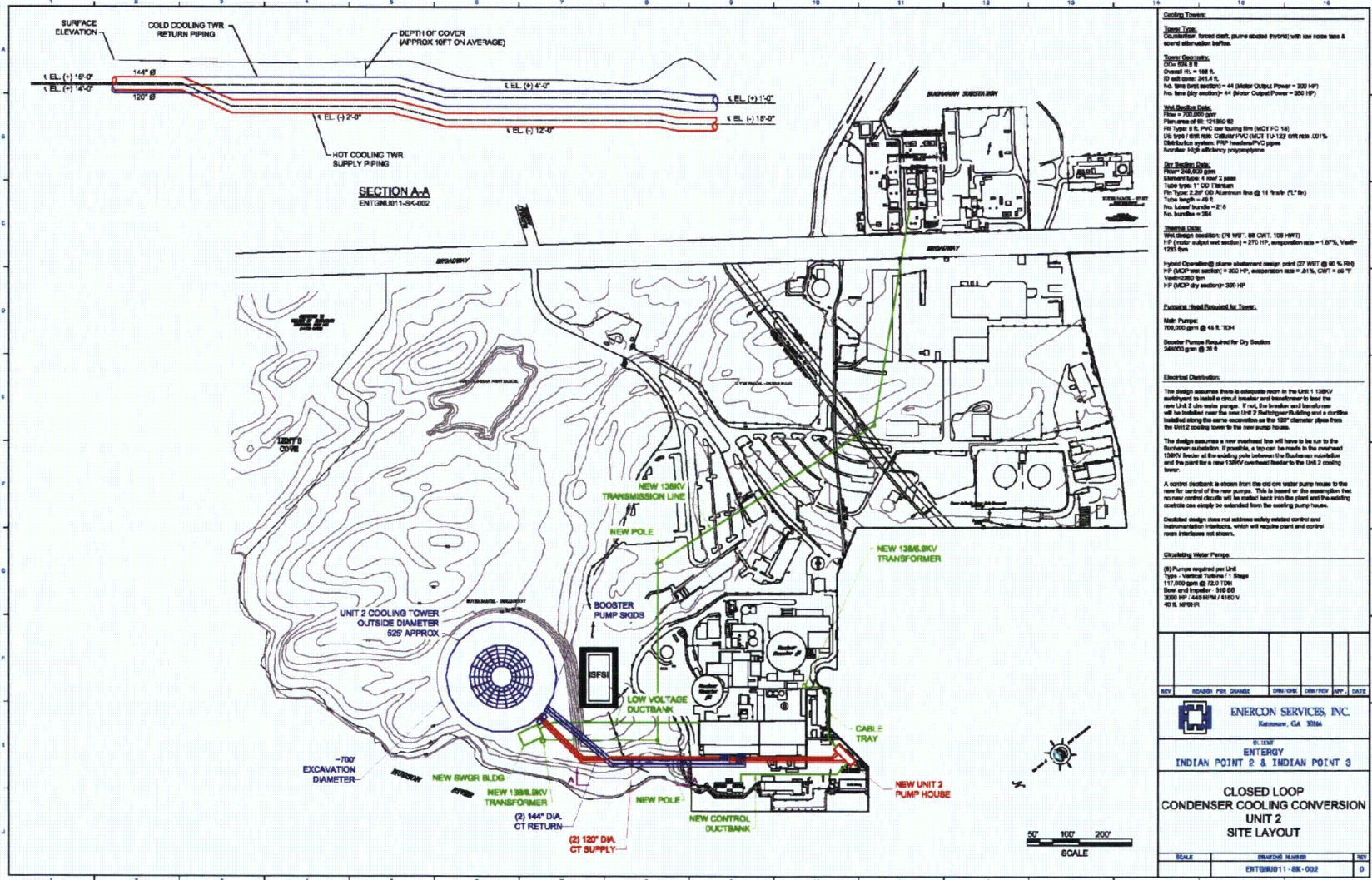
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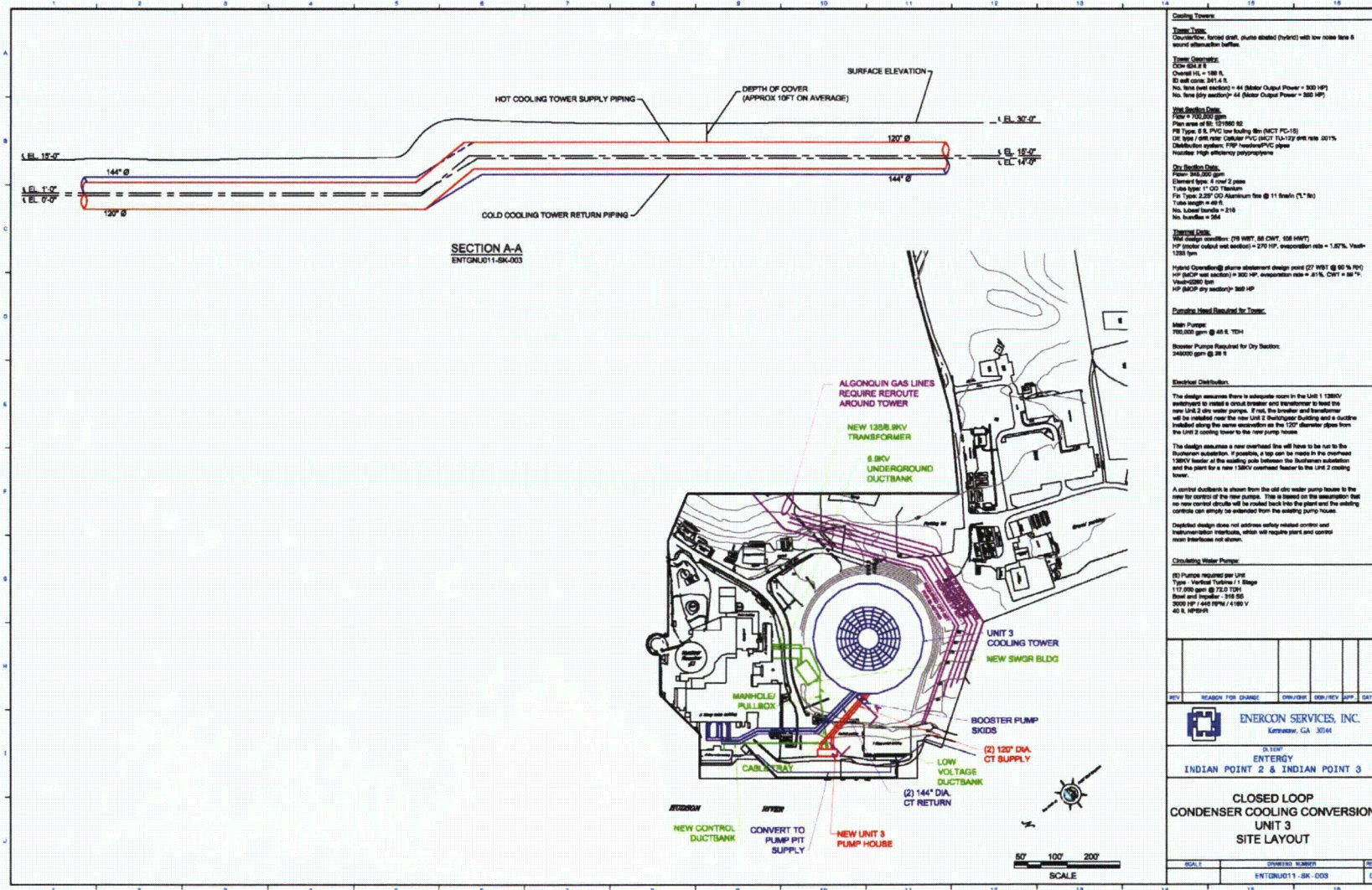
**ENERCON SERVICES, INC.**  
 Kennesaw, GA 30144

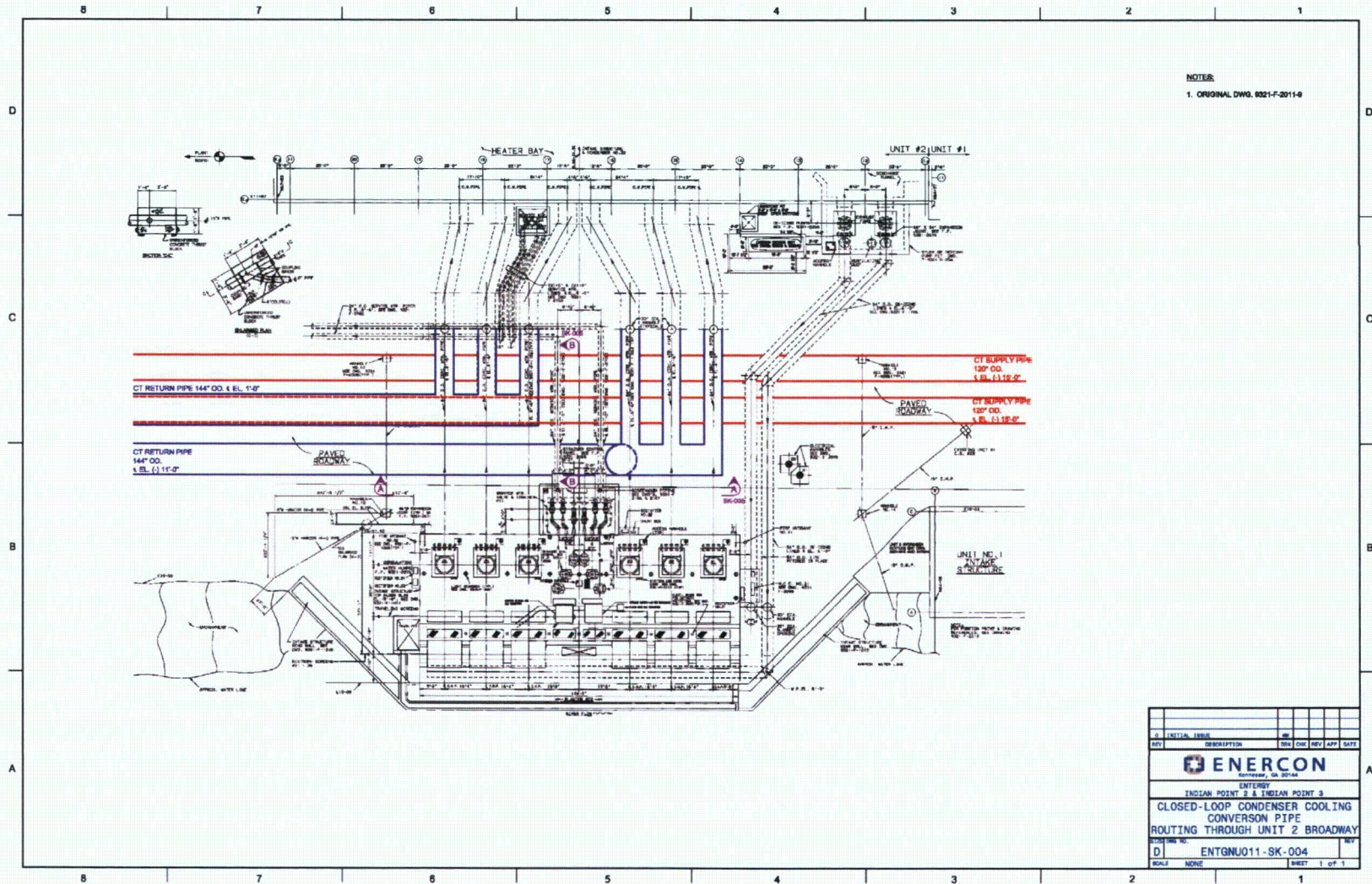
3-10187  
 ENERTECH  
 INDIAN POINT 2 & INDIAN POINT 3

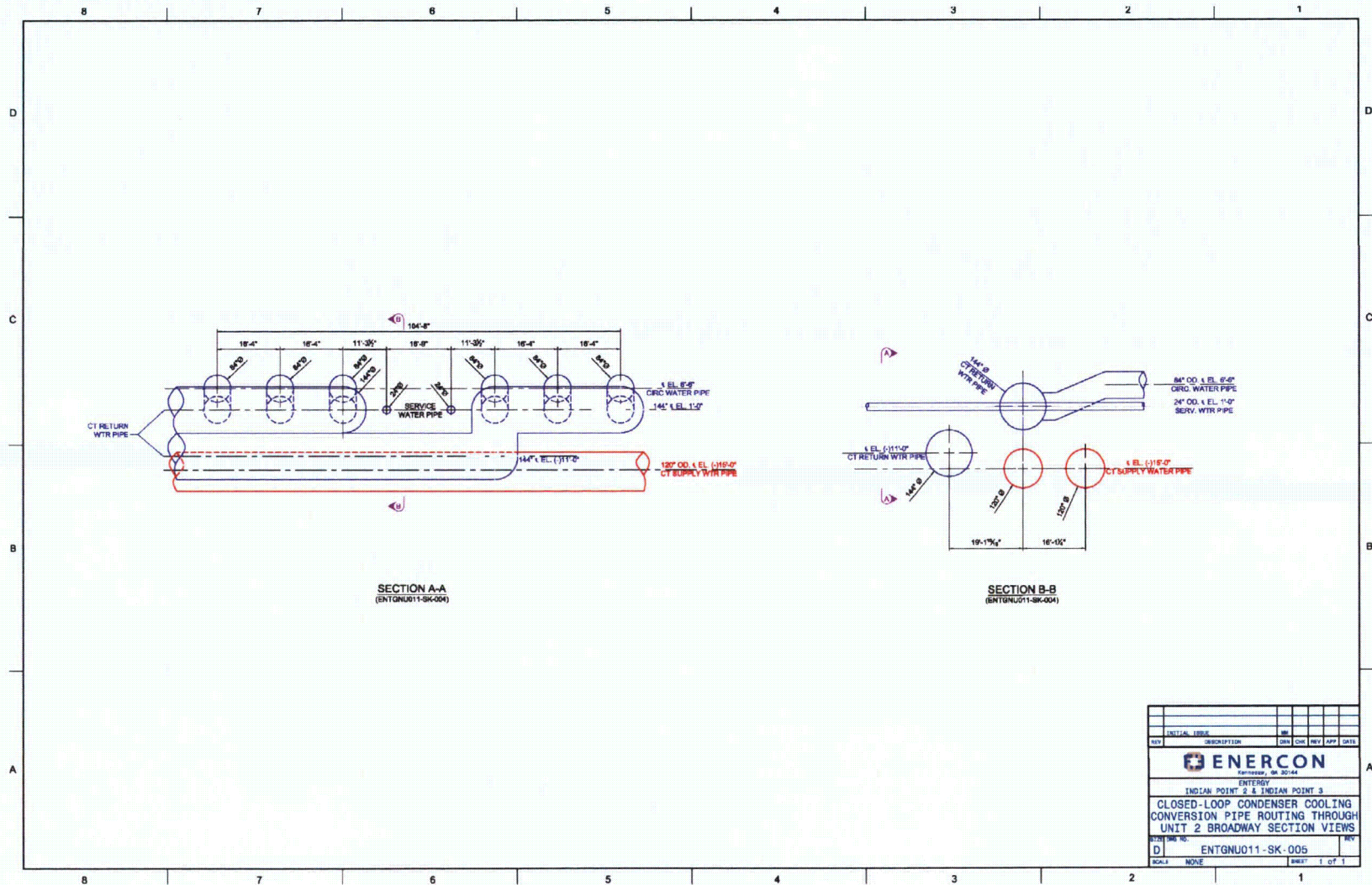
**CLOSED LOOP  
 CONDENSER COOLING CONVERSION  
 SITE LAYOUT**

SCALE	DRAWING NUMBER	REV
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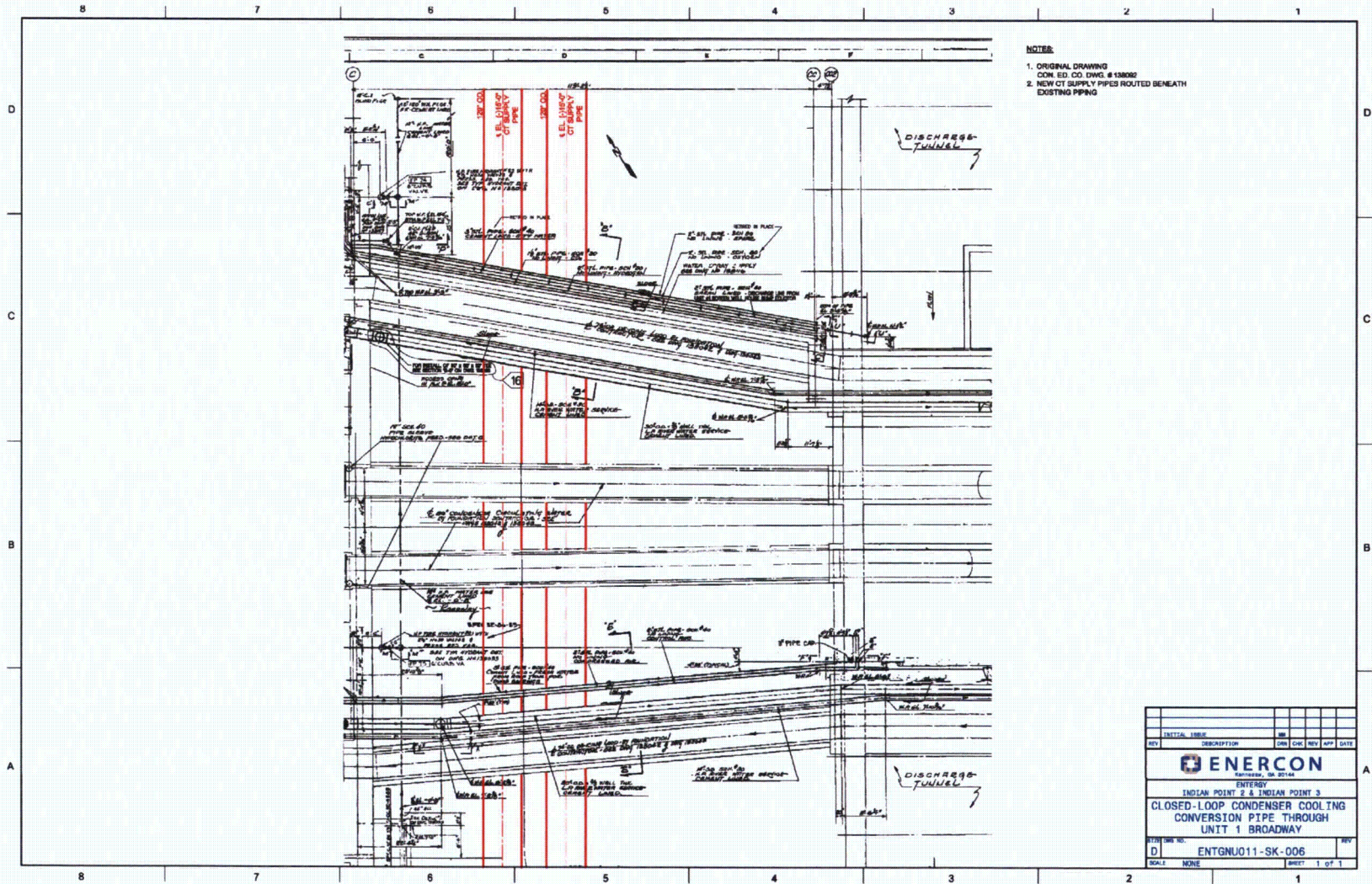






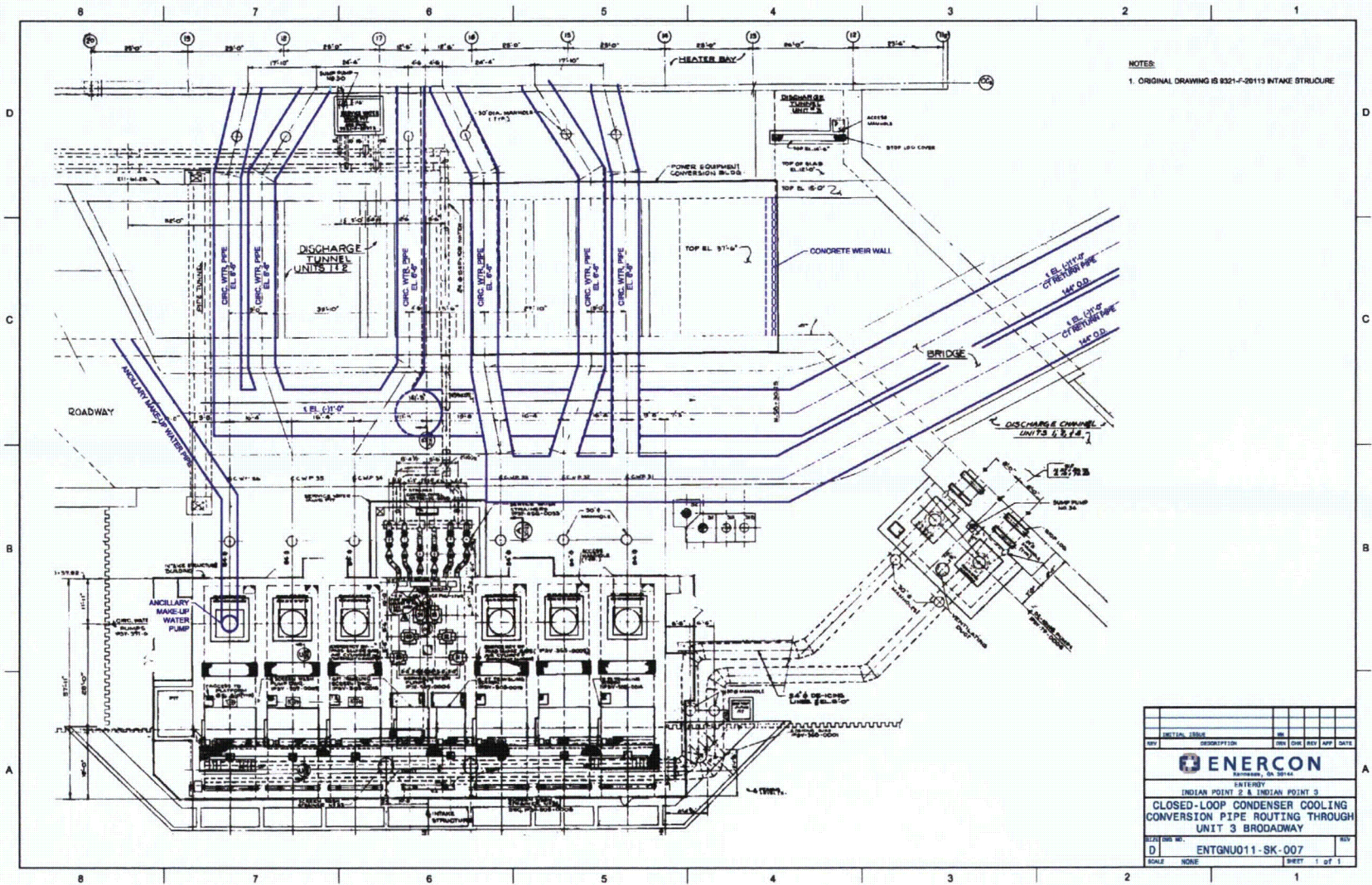
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<b>ENERCON</b>					
ENTGNDU11-SK-004					
INDIAN POINT 2 & INDIAN POINT 3					
CLOSED-LOOP CONDENSER COOLING CONVERSION PIPE ROUTING THROUGH UNIT 2 BROADWAY SECTION VIEWS					
SIZE	REV				
D	ENTGNDU11-SK-005				
SCALE	NONE				
					SHEET 1 of 1





- NOTES:**
1. ORIGINAL DRAWING  
CON. ESD. DD. DWG. # 130002
  2. NEW CT SUPPLY PIPES ROUTED BENEATH EXISTING PIPING

INITIAL	ISSUE	DATE	BY	CHK	REV	APP	DATE
ENTERGY INDIAN POINT 2 & INDIAN POINT 3							
CLOSED-LOOP CONDENSER COOLING CONVERSION PIPE THROUGH UNIT 1 BROADWAY							
DATE	ISSUE NO.	REV					
D	ENTGNOJ011-SK-006						
SCALE	NONE	SHEET 1 OF 1					



**Attachment 3**

**Subsurface Radiological Considerations Related to Construction of  
Closed-Loop Cooling at Indian Point Energy Center Units 2 and 3**

**GZA GeoEnvironmental, Inc.**



GZA  
GeoEnvironmental, Inc.

Engineers and  
Scientists



New York  
104 West 29<sup>th</sup> Street  
10<sup>th</sup> Floor  
New York, NY 10001  
Phone: 212-594-8140  
Fax: 212-279-8180

Connecticut  
120 W Grove Street  
Bloomfield, CT 06824  
Phone: 860-243-9055  
Fax: 860-243-9055

Massachusetts  
One Edgewater Drive  
Norwood, MA 02062  
Phone: 781-278-3700  
Fax: 781-278-5701

**SUBSURFACE RADIOLOGICAL CONSIDERATIONS  
RELATED TO CONSTRUCTION OF CLOSED-LOOP COOLING  
AT INDIAN POINT ENERGY CENTER UNITS 2 AND 3**

**Introduction**

As noted in the Enercon Services, Inc. ("Enercon") closed-loop cooling report<sup>1</sup> (the "Enercon Report") and proposed by the New York State Department of Environmental Conservation ("NYSDEC"), construction of two large counter-flow, forced draft, plume abated hybrid cooling towers and associated piping (the "NYSDEC Proposed Project") at Indian Point Energy Center ("IPEC") Units 2 and 3, requires excavation and disposal of soil and bedrock at the basins for the proposed Unit 2 and Unit 3 cooling towers, as well as for the piping trenches that connect the proposed cooling towers to their respective circulating water systems. Entergy Nuclear Indian Point 2, LLC and Entergy Nuclear Indian Point 3, LLC ("Entergy") retained GZA GeoEnvironmental, Inc. ("GZA") to address radiological contamination issues as part of its feasibility assessment undertaken pursuant to the NYSDEC Assistant Commissioner's August 13, 2008 Interim Decision. GZA has extensive experience monitoring the groundwater on the IPEC site using various instrumentation installations which were constructed as part of the 2008 Hydrogeologic Site Investigation.<sup>2</sup>

As can be seen on Figures 1 and 2, groundwater containing Tritium and Strontium migrates through a portion of the excavation area for the NYSDEC Proposed Project. The delineated (shaded) Tritium and Strontium "plume areas" are defined as areas on these figures where the rolling yearly average groundwater radionuclide activities<sup>3</sup> are greater than 5,000 pCi/L and 2

<sup>1</sup> Economic and Environmental Impacts Associated with Conversion of Indian Point Units 2 and 3 to a Closed-Loop Condenser Cooling Water Configuration, by Enercon Services, Inc.

<sup>2</sup> Hydrogeologic Site Investigation Report, January 7, 2008, prepared by GZA GeoEnvironmental, Inc., on behalf of Enercon Services, Inc., for Entergy Nuclear Northeast, Indian Point Energy Center, 450 Broadway, Buchanan, NY 10511.

<sup>3</sup> For radionuclides in groundwater, the level of contamination is reported as the activity of each radionuclide, which is the measured radiation intensity emitted by each radionuclide per unit time in a specified volume of water (e.g., picocurie per liter – pCi/L).

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pCi/L, respectively.<sup>4</sup> GZA previously developed the plume delineations and rolling average data shown on Figures 1 and 2 based on quarterly groundwater sampling and elevation measurements taken at various monitoring points during 2008, as well as the analyses summarized in previous Quarterly Groundwater Monitoring Reports<sup>5</sup> and the Conceptual Site Model ("CSM") presented in the 2008 Hydrogeologic Site Investigation Report.<sup>6</sup> The Figures 1 and 2 also present radionuclide activities for each individual sampling depth at each location, both within and outside of the delineated (shaded) plume areas. In addition to Tritium and Strontium, other radionuclides such as Cesium, Cobalt and Nickel (additional primary indicator radionuclides) have also been detected on the site. These detections are generally disperse, however, and do not appear to represent defined groundwater plumes. The more recent data for Cesium, Cobalt and Nickel are presented in the Quarterly Groundwater Monitoring Reports, and the more historic data are presented in the Hydrologic Site Investigation Report.<sup>7</sup>

As discussed in further detail below, groundwater exists in the interstitial pore spaces and fractures in bulk soil and bedrock deposits. If groundwater within the soil and bedrock is contaminated, excavated material will also be contaminated, and sampling and analysis protocols would be developed and employed. Moreover, construction of the NYSDEC Proposed Project would require "dewatering," a process used to maintain dry conditions during construction. Due to site conditions, dewatering would involve using continuous pumping of contaminated groundwater. As also discussed in further detail below, this dewatering would require special precautions to prevent migration of radionuclide contaminated groundwater to clean areas.

**Locations For Which Sampling and Analysis Protocols Would be Developed and Employed for Construction of the NYSDEC Proposed Project**

GZA understands from Entergy's counsel that NRC regulations (10 C.F.R. Part 20 Subpart K) require that all radionuclide contaminated soil and bedrock excavated for construction of the NYSDEC Proposed Project be identified, handled, treated and disposed of appropriately. Limited radionuclide data exists for soil and bedrock in the areas requiring excavation under the NYSDEC Proposed Project, particularly at depth. Therefore, GZA used the areal and vertical extent of radionuclide contamination in groundwater as an indicator of soil and bedrock contamination for excavation management planning purposes. Specifically, if groundwater within the bulk soil/bedrock deposit is contaminated, the excavated bulk material will also be contaminated (in part, because contaminated groundwater will be retained in the bulk material during and after

<sup>4</sup> The plume delineation boundary values were established at one-quarter of the drinking water standards for these radionuclides. Although GZA emphasizes that drinking water standards (USEPA MCLs) do not apply to the IPEC property given that there are no drinking water sources on or proximate to the site, the MCLs do provide a useful benchmark for comparisons of relative human risk. Where yearly rolling average radionuclide activity data were available for multiple depths at a given location, GZA used the highest value to develop plume delineations. This is a typical approach to represent three-dimensional contaminant data sets on two-dimensional maps.

<sup>5</sup> Quarterly Groundwater Monitoring Reports are required under IP Site Management Manual, Quality Related Administrative Procedure, IP-SSM, CY-110.

<sup>6</sup> Hydrogeologic Site Investigation Report, January 7, 2008, prepared by GZA GeoEnvironmental, Inc. on behalf of Enercon Services, Inc., for Entergy Nuclear Northeast, Indian Point Energy Center, 450 Broadway, Buchanan, NY 10511.

<sup>7</sup> GZA notes that Cesium, Cobalt and Nickel have not been detected in the groundwater within the proposed excavation areas during the more recent sampling for the Quarterly Groundwater Monitoring Reports (Quarter 2 2007 through Quarter 4 2008). However, GZA has previously detected Cesium and Nickel in groundwater within these areas, as well as Nickel in areas upgradient of the proposed excavations for the NYSDEC Proposed Project.

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excavation). Groundwater contaminant levels (in pCi/L), however, are not directly translatable into soil/bedrock activity levels (in pico curie per kilogram - pCi/Kg), particularly for radionuclides such as Strontium, which preferentially partition out of the groundwater and adsorb onto the solid mineral surfaces of the soil grains/intact bedrock.<sup>8</sup> Thus, it is GZA's recommendation that sampling and analysis protocols would be developed and employed to assess actual radionuclide levels in the excavated material. These sampling and analysis protocols would be developed to properly and prudently identify potentially contaminated soil/bedrock within the proposed excavation areas.

Within the shaded groundwater plumes of Figures 1 and 2, and at other identified locations known to contain radionuclides elevated above Minimum Detectable Concentrations ("MDCs"),<sup>9</sup> it is GZA's recommendation that sampling and analysis protocols would be developed and employed to prudently determine appropriate handling and disposal requirements and manage the work in accordance with NRC regulations. In addition, sampling and analysis protocols would also be employed in areas outside, but proximate to the delineated plumes and known areas of contamination. This prudent course of action is considered appropriate to manage the risks always inherent in subsurface work, and in particular because of the site's subsurface hydrogeologic variability. Moreover, it is GZA's opinion that sampling and analysis protocols would also be needed because Entergy's site acquisition is recent relative to the extended extent of site history.

Based on the existing groundwater data, additional areas which would undergo testing for construction of the NYSDEC Proposed Project for Unit 2 extend northward from the shaded plume at least to the northern edge of the Riverfront area, and southward from the shaded plume to the proposed Unit 2 pump house structure (see Figures 1 and 2). In addition, although there is no plume delineated on these figures in the Unit 3 Riverfront area, radionuclide activity is expected in the groundwater in this area, but below the plume delineation limits (enumerated above). As such, additional sampling and analysis protocols also would be applied to material excavated from the Unit 3 Riverfront area (at least from the proposed Unit 2 pump house structure to the North, to the proposed Unit 3 pump house structure to the South) to increase the level of certainty that all radionuclide contaminated material requiring special handling and disposal would be identified prior to disposal.

#### Construction Dewatering

During the excavation process, not only would soil and bedrock contaminated with radionuclides require disposal, but contaminated groundwater located under the site would also need to be continually pumped from the excavation areas to maintain dry conditions required for construction and backfilling. This is because the NYSDEC Proposed Project would require excavations at depths well below the groundwater table. As advised by Entergy's consultant Enercon, excavations would begin at the cooling tower basin locations, followed by excavation of proximate pipe trenching, with pipe trenching in the Riverfront area (the primary location of

<sup>8</sup> In these cases, the radionuclides may become concentrated on the solid surfaces over time. The degree to which this time-dependent concentration process occurs is highly site specific (both from site to site and within a single site), and therefore is difficult to predict and/or establish experimentally.

<sup>9</sup> The groundwater is also contaminated with radionuclides in some areas outside of the shaded plumes, but at activities below the plume delineation levels of 5,000 pCi/L and 2 pCi/L (for Tritium and Strontium respectively). Where currently detected above MDC, these data have been provided on Figures 1 and 2. For a more complete summary of these detections over time, refer to the Quarterly Groundwater Monitoring Reports and the Hydrogeologic Site Investigation Report.



the identified contaminated groundwater) completed last. This excavation sequencing would be required to minimize plant outage time during construction. If construction dewatering was implemented in the same sequence as excavation (as would be typical construction procedure<sup>10</sup>) however, it would cause the groundwater to migrate from contaminated areas to clean areas,<sup>11</sup> thus resulting in spreading of radionuclides in the subsurface. To control this "plume spreading," an atypical dewatering sequence would have to be adopted. This would entail starting the groundwater dewatering in the contaminated areas (within Riverfront in the vicinity of Unit 2 and Unit 1, and potentially Unit 3<sup>12</sup>) prior to excavation in these areas, but coincident with excavation and dewatering in the adjacent clean areas which would be excavated first.<sup>13</sup> This dewatering would also have to be continued until completion of the excavation. As such, dewatering would be executed for a longer period of time than would be typical of normal construction practice, thus increasing costs.

It is anticipated that the dewatering would be accomplished with a line of closely spaced groundwater extraction wells (on the order of 25 to 75 feet apart) on each side of the trench excavation in the Riverfront area. These wells would likely extend to an elevation of approximately (-) 45 feet so as to allow the groundwater elevation to be reduced to approximately elevation (-) 20 feet (to below the bottom of the excavation) along the centerline of the pipe trench. Based on the 3-dimensional numeric computer model previously developed by GZA as part of the 2008 Hydrogeologic Investigation for the IPEC site, dewatering flow rates from the contaminated plume area in the vicinity of the Unit 2 proposed piping trenches are estimated to be approximately 60 gallons per minute (gpm) prior to excavation and dewatering in adjacent areas. As advised by Enercon, it is expected that this phase of the dewatering would last approximately one month, resulting in the collection of approximately 2.7 million gallons of radionuclide contaminated groundwater. Once dewatering begins in the adjacent clean areas, the flow rate from the contaminated area is anticipated to decrease to approximately 55 gpm.<sup>14</sup> As further advised by Enercon, it is expected that this phase of the

<sup>10</sup> Typically, construction sequencing is developed to minimize the area and length of time over which dewatering is employed to reduce its added cost. As such, dewatering is typically not started until the excavation is about to reach the groundwater depth, and is only employed in the area actually being excavated. Therefore, dewatering would start in the cooling tower basin locations and would continue there until construction was completed up to the water table surface, and then it would be discontinued in this area. As the excavation proceeded from the basin locations along the piping trenches, the dewatering would just precede the excavations as they advanced towards, and then into the contaminated Riverfront area.

<sup>11</sup> With typical dewatering sequencing, as the excavation, and thus the dewatering, approaches the plume within Riverfront area, the groundwater elevation in the dewatered area would be below that in the plume area. The contaminated groundwater at the higher elevation would thus flow downward towards the lower elevations in the dewatered area, thus causing the radionuclide plume to migrate into clean areas where it had not previously existed.

<sup>12</sup> The dewatering flow rates subsequently provided below apply only to those areas involving contaminated groundwater from the Tritium and Strontium plumes in the Unit 2 proposed piping trench area; plumes shown as shaded in Figures 1 and 2. It is predicted that there is also Tritium and Strontium in the groundwater in the Riverfront area down gradient of Unit 3, but at concentrations below approximately 3000 pCi/L and 1.5 pCi/L, respectively. Although lower than the levels within the shaded plumes, these activities are still above those predicted in the area of the Unit 3 cooling tower and the associated piping. As such, dewatering for the Unit 3 piping trenches may require similar constraints, and thus added cost, for dewatering sequencing, treatment and disposal similar to that for the Unit 2 area.

<sup>13</sup> With dewatering in the contaminated area preceding that in the adjacent clean areas, the groundwater elevations in the contaminated area could be maintained below those in the clean areas. Therefore the clean water would then be at the higher elevations and would thus flow downward into the contaminated area. Clean groundwater flowing in would then prevent the contaminated groundwater from flowing out, thus preventing the plume from spreading into cleaner areas.

<sup>14</sup> Prior to the start of dewatering in the adjacent clean areas, clean water to the North and South of the contaminated area would be pulled into the contaminated area by the dewatering wells, mixed with the contaminated water and pumped out for disposal. Once dewatering starts in the adjacent clean areas, this

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work will require 12 months, resulting in the collection of approximately 29 million additional gallons of radionuclide contaminated groundwater (approximately 32 million Gallons total).<sup>15</sup> As indicated above, the dewatering flow rates and volume estimates provided above are for the Unit 2 proposed piping trenches only; additional dewatering constraints, and the associated cost increases, may need to be implemented for the Unit 3 proposed piping trenches to control the potential spreading of lower level radionuclides. If dewatering were to be required to control plume spreading in this area also, it would more than double the flow rate requiring treatment and disposal.<sup>16</sup>

#### Disposal of Radionuclide Contaminated Water Which Would be Generated During Construction Dewatering

Once groundwater is extracted by the dewatering systems, it must be disposed of in accordance with applicable rules and regulations. Groundwater extracted from the contaminated area will contain Tritium, Strontium and potentially other radionuclides at lower activity levels, including Cesium, Cobalt and Nickel. Treating the extracted groundwater prior to disposal increases the range of possible disposal options. Treatment methods exist for removing radionuclides from water, except for Tritium. A primary treatment method is to flow the contaminated water through various beds of specialized resins, which remove the radionuclides. Based on Enercon and Entergy evaluation of ongoing treatment of groundwater at Unit 1, it is anticipated that this treatment would cost at least \$ 7,000,000 for Operation and Maintenance, and disposal of contaminated treatment resin. Even this treatment, however, still leaves Tritium in the extracted water, for which there is no practical removal method.<sup>17</sup> Therefore, the groundwater extracted during piping construction in and proximate to the contaminated areas will contain Tritium after treatment, potentially impacting, and limiting, disposal options.

The most practical and cost effective method to dispose of the Tritium-contaminated groundwater generated during the dewatering of excavations for the NYSDEC Proposed Project would be to dispose of it to the Hudson River (with prior dilution through the Discharge Canal) in compliance with NRC regulations. However, Entergy's counsel has advised GZA that NYSDEC appears to believe they have jurisdiction over disposal of groundwater contaminated with radionuclides.<sup>18</sup> It is GZA's understanding that NYSDEC further believes disposal of radionuclide contaminated groundwater to the Hudson River is not permissible under State regulations. If disposal of the Tritiated groundwater to the Discharge Canal is thus found to be infeasible, three other potential disposal options exist, but each has serious drawbacks which would likely render it technically infeasible, as discussed below.

flow from the North and South will be picked up by these additional clean area dewatering wells, and will remain uncontaminated. Therefore, the flow of water into the contaminated area will decrease somewhat.

<sup>15</sup> During this portion of the work, substantial additional groundwater will also be generated by dewatering in areas outside the contaminated areas (primarily associated with dewatering for the cooling tower basin excavations). The volume of this dewatering flow has not been quantified herein given that it should not be contaminated and thus would not require special treatment and disposal. Therefore, it is assumed that this water can be handled as part of the typical construction process and discharged to the Hudson River without treatment for radionuclides.

<sup>16</sup> Based on the 3-dimensional numeric computer model previously developed by GZA as part of the 2008 Hydrogeologic Investigation for the IPEC site, indicates a projected dewatering flow rate for the Unit 3 area would be approximately 100 gpm, as compared to that for the Unit 2 area of 60 gpm.

<sup>17</sup> Methods do exist to remove tritium from water. However, these methods are not cost effective for the high groundwater flows required to dewater the excavations for the NYSDEC Proposed Project.

<sup>18</sup> NYSDEC letter of May 13, 2009 regarding: Notice of Incomplete Application / Request for Additional Information, Joint Application for CWA § 401 Water Quality Certification, NRC License Renewal – Indian Point Units 2 and 3, DEC Nos:3-5522-00011/00030 [IP2] and 3-5522-00105/00031 [IP3].





The first potential option would be to pump the water to the sanitary sewer system. It appears that this option would be in compliance with NRC regulations (10 C.F.R. Part 20 Subpart K § 20.2003, Disposal by Release into Sanitary Sewerage). Sanitary sewer disposal is also routinely utilized for non-radiological contamination, such as industrial chemical discharges, because the municipal treatment systems can eliminate many of these chemical contaminants prior to ultimate discharge to a surface water body. However, there are a number of issues which must be addressed to demonstrate the feasibility of this option. First, the IPEC sanitary lift station pumping capacity, as well as the piping to the sanitary sewer connection may need to be upgraded to handle the increased flow (the increase in flow from dewatering would more than triple the allowed normal average daily flow). Second, even if the IPEC system could handle this discharge, intentional discharge of groundwater to the sanitary system, whether contaminated or not, is specifically disallowed by the Agreement with the Village of Buchanan.<sup>19</sup> Third, even if the Village agreed to modify the terms of the current Agreement, it is unlikely that the Publically Owned Treatment Works ("POTW") is physically capable of handling the added dewatering flow rate. As provided by Enercon, the Buchanan POTW has a listed total capacity of 0.5 million gallons per day ("mgd"), which is equal to approximately 350 gpm. The dewatering flow rate for the plume area of the Unit 2 proposed piping trenching is 55 to 60 gpm. This is over 15% of the total POTW capacity absent any contingency for the potential inclusion of Unit 3 proposed piping trench dewatering flow (a projected additional 100 gpm, bringing the potential total additional flow to nearly 50% of the total plant capacity). Therefore, it does not appear that disposal of the groundwater dewatering flow to the POTW is technically feasible, even if the Village of Buchanan were willing to amend the Agreement.<sup>20</sup>

The second potential option would be to truck the water off-site for disposal. As advised by Enercon, there are only two operating facilities in the country, Energy Solutions and Perma-Fix, which can accept and dispose of free liquids contaminated with radionuclides, provided that they at least meet Class A waste acceptance criteria.<sup>21</sup> Off-site disposal would likely entail trucking through local roads. Each truck can carry 6000 gallons of water,<sup>22</sup> amounting to over 10,500 total truck trips. (one trip in to pick up and one trip out for disposal for each 6,000 gallons of water). During peak pumping periods, it is expected that the frequency would be approximately 30 truck trips per day, on both week days and weekends. The cost for this off-site disposal option would be approximately \$10 / gallon for transportation and disposal, for a total cost of \$320 million.<sup>23</sup> However, this option was found to be infeasible, beyond just the costs involved,

<sup>19</sup> "Agreement between Village of Buchanan and IPEC", dated July 1, 1987

<sup>20</sup> It is noted that the Town of Peekskill POTW has a larger capacity of 10mgd, as based on data provided by Enercon. However, this POTW is 4 miles away, rendering its use impractical.

<sup>21</sup> 10 C.F.R. 61.55 Code of Federal Regulations, Title 10, Energy, Part 61, "Licensing Requirements for Land Disposal of Radioactive Waste."

<sup>22</sup> Large tanker trucks range in capacity from 5,500 gallons to 9,000 gallons. A common truck size of 6,000 gallons was selected for this computation to balance a reduced number of truck trips (as compared with 5,500 gallon or smaller capacity trucks), with the difficulties associated with larger trucks (e.g., 9,000 gallon capacity) on smaller local roads.

<sup>23</sup> These values account for the Unit 2 piping trenches only; additional dewatering constraints, and the associated truck trip and cost increases, may need to be implemented for the Unit 3 proposed piping trenches.

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given the dewatering pumping rate of 55 to 60 gallons per minute (80,000 to 90,000 gallons per day) far exceeds the capacity of this disposal facility.<sup>24</sup>



The third potential option would be to attempt to re-inject the extracted groundwater back into the subsurface on-site. This could potentially be accomplished with a series of re-injection wells drilled into the bedrock along the upgradient portion of the plume area. In concept, the contaminated water would be kept in continual recirculation within the existing on-site plume. In reality, however, this could cause plume spreading at the IPEC site. To begin with, it is generally much easier to extract groundwater from the subsurface than it is to re-inject this groundwater back into the same subsurface deposits<sup>25</sup>. This general "rule of thumb" is particularly true for fractured bedrock, such as exists at the IPEC site. The added difficulty expected for re-injection would likely therefore spread Tritiated water outside the boundaries of the existing plume. This would not only then cause plume spreading, but the spreading could easily be along bedrock fractures of indeterminate location and extent (on and potentially off-site). In addition, the area of excavation that would require dewatering for piping construction is located relatively close to the Hudson River. As such, sufficient extraction to prevent plume spreading in this area would inevitably induce some infiltration of river water back into the site, and into the extraction wells. This clean water would mix with the contaminated water, resulting in a greater volume of contaminated water than that currently flowing within the plume.<sup>26</sup> Therefore, re-injection of this larger volume of contaminated water at the upgradient edge of the plume would necessarily result in plume spreading; i.e., the plume would have to expand to carry the additional flow. Given the above analysis, on-site reinjection of the contaminated water extracted during construction dewatering for the NYSDEC Proposed Project is considered infeasible because of the on-site, and potentially off-site, plume spreading it would inevitably cause.

### Conclusions

It is GZA's recommendation that excavation for the NYSDEC Proposed Project would employ implementation of sampling and analysis protocols for the soil and bedrock removed from the piping trenches along the entire Riverfront area, from its most northern extent to the proposed Unit 3 Pump House to the South.

In addition, the dewatering used to maintain dry construction conditions within the proposed piping trenches would require atypical construction sequencing to prevent the spreading of radionuclide contaminated groundwater into cleaner areas. This required control of plume spreading would result in more prolonged and extensive dewatering than would otherwise be required if the groundwater was not contaminated, thus increasing operation and treatment costs.

<sup>24</sup> As advised by Enercon, Energy Solutions and Perma-Fix, both located in Oak Ridge, TN., can currently accept only a total of 3600 gallons of free liquid containing radionuclides per day. It is assumed that a portion of this total capacity would be utilized by others at the time of construction dewatering, thereby reducing the capacity available for accepting the IPEC dewatering flow. Even at full capacity, this facility could only treat about 4 percent of the projected dewatering flow rate from the Unit 2 area only (absent any contingency for the potential inclusion of an additional flow of nearly 150,000 gpd for Unit 3 piping trench dewatering).

<sup>25</sup> F. G. Driscoll, "Groundwater and Wells"; Johnson Division; 1986. St. Paul, MN; pg. 771

<sup>26</sup> In addition, other sources of clean water would be captured during dewatering of the plume area, and would further increase the volume of the water requiring re-injection. These include the clean groundwater at the periphery of the plume to the North and South at the beginning of dewatering (as described above) and infiltration of precipitation.



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After treatment, the radionuclide contaminated groundwater, which would still contain Tritium, must be disposed of. Given the continuous flow rate of 55 to 60 gpm, practical disposal options appear limited to discharge to the Hudson River. While disposal to the Hudson River appears consistent with NRC regulations, GZA has been advised by Entergy's counsel that NYSDEC appears to believe they have jurisdiction over disposal of groundwater contaminated with radionuclides, and believes that disposal to the Hudson River is not permissible under State regulations. GZA concludes, therefore, that disposal options for radionuclide contaminated groundwater generated by the NYSDEC Proposed Project are limited.

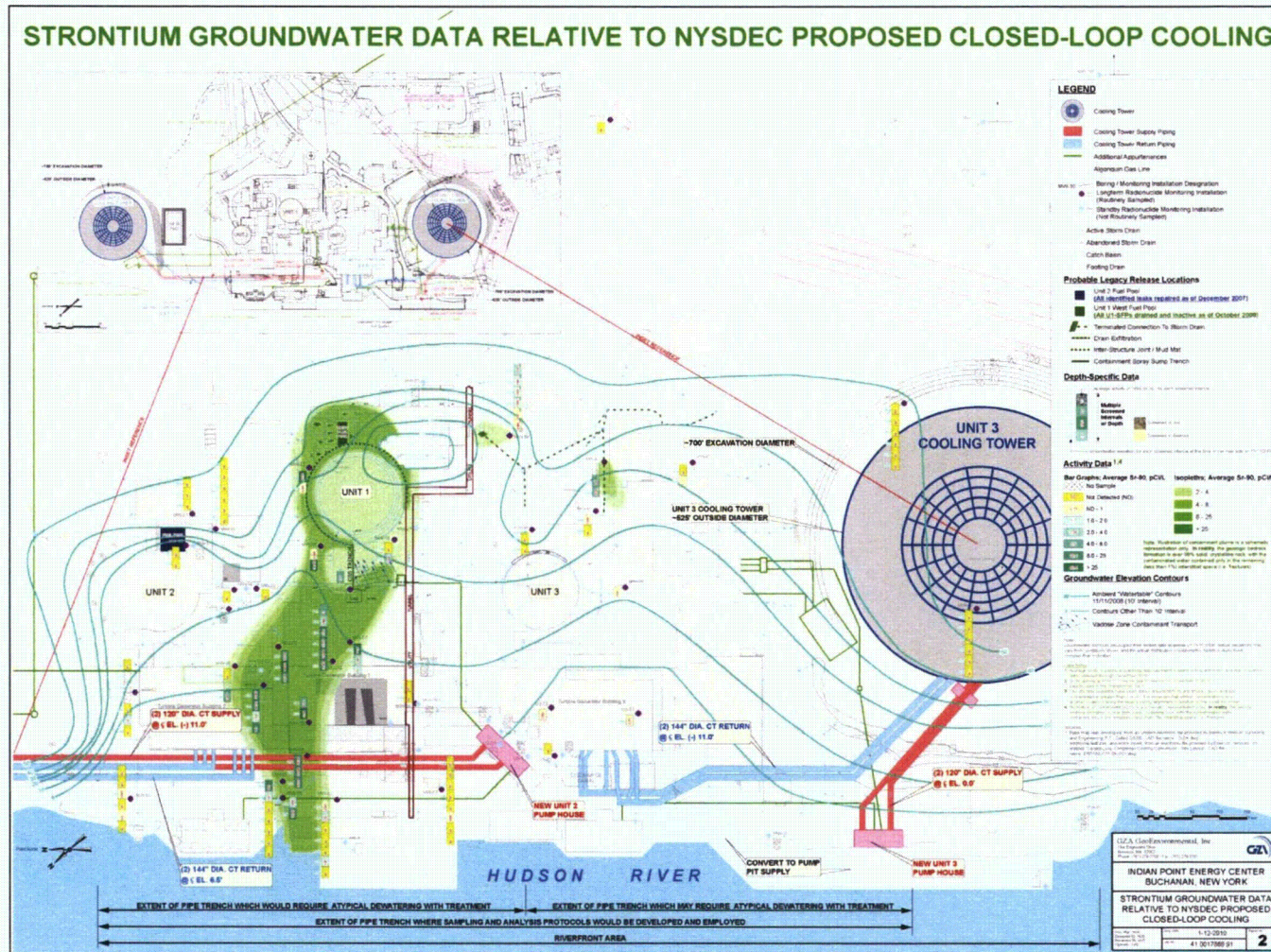


Figure 1 Conceptual Closed-Loop Cooling Layout with Strontium Groundwater Data

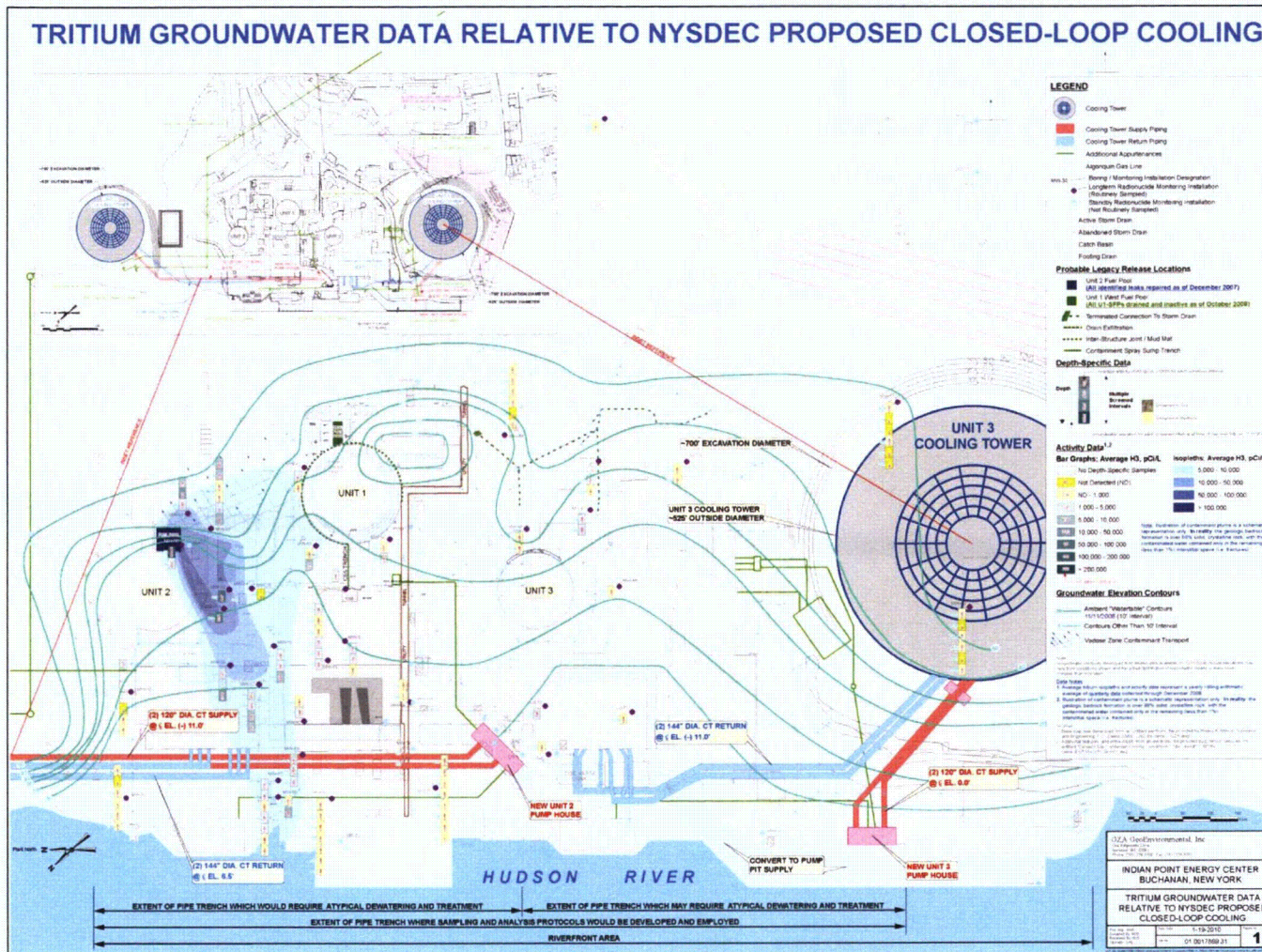


Figure 2 Conceptual Closed-Loop Cooling Layout with Tritium Groundwater Data

**Attachment 4**  
**Closed-Loop Performance**

- Section 1: Data Recovery Rates**
- Section 2: PEPSE Models for IPEC Units 2 and 3**
- Section 3: Annual Operational Losses for IPEC Units 2 and 3**
- Section 4: Monthly Operational Losses for IPEC Unit 2**
- Section 5: Monthly Operational Losses for IPEC Unit 3**

Table 4-1 IPEC Meteorological Data Recovery Rate (2001-2008)

Month	Recover Rate (%)
January	99.8%
February	99.8%
March	99.0%
April	99.6%
May	99.2%
June	99.5%
July	99.7%
August	90.8%
September	86.8%
October	92.4%
November	99.9%
December	99.6%
Annual	97.2%

<sup>1</sup> Meteorological data recovery rate is significantly better than is typical seen in the commercial nuclear power industry, and far exceeds the 90% recovery rate criteria set forth by RG 1.23 (Ref. 12.27)

Table 4-2 Hudson River Temperature Data Recovery Rate (2001-2008)

Month	Recover Rate (%)
January	100.0%
February	100.0%
March	99.6%
April	100.0%
May	99.6%
June	100.0%
July	100.0%
August	99.6%
September	99.2%
October	100.0%
November	100.0%
December	100.0%
Annual	99.8%

Performance evaluation of power system efficiency (PEPSE) is a steady-state energy balance software program that calculates the performance of electric generating plants. It is used throughout the world by fossil-fired plants, nuclear plants, gas turbine plants, combined cycles plants, and plants with atypical fluid systems. A plant analysis model is constructed by the user's development of a plant schematic that mimics the actual plant component connections. The plant analysis model is benchmarked against actual plant data to ensure accuracy of the analysis.

The evaluation of closed-loop cooling for IPEC is performed using a PEPSE model that provides the expected plant operational parameters and power reductions associated with conversion of IPEC to closed-loop cooling. The PEPSE model is tuned to each Unit's actual performance and, therefore, provides an accurate summary of the expected results of conversion to closed-loop cooling.



The IPEC PEPSE models were reviewed, updated, and run. The figures below show the IPEC PEPSE models, input similar to a general arrangement drawing, after they have been updated and run at a user defined circulating water temperature of 65°F.

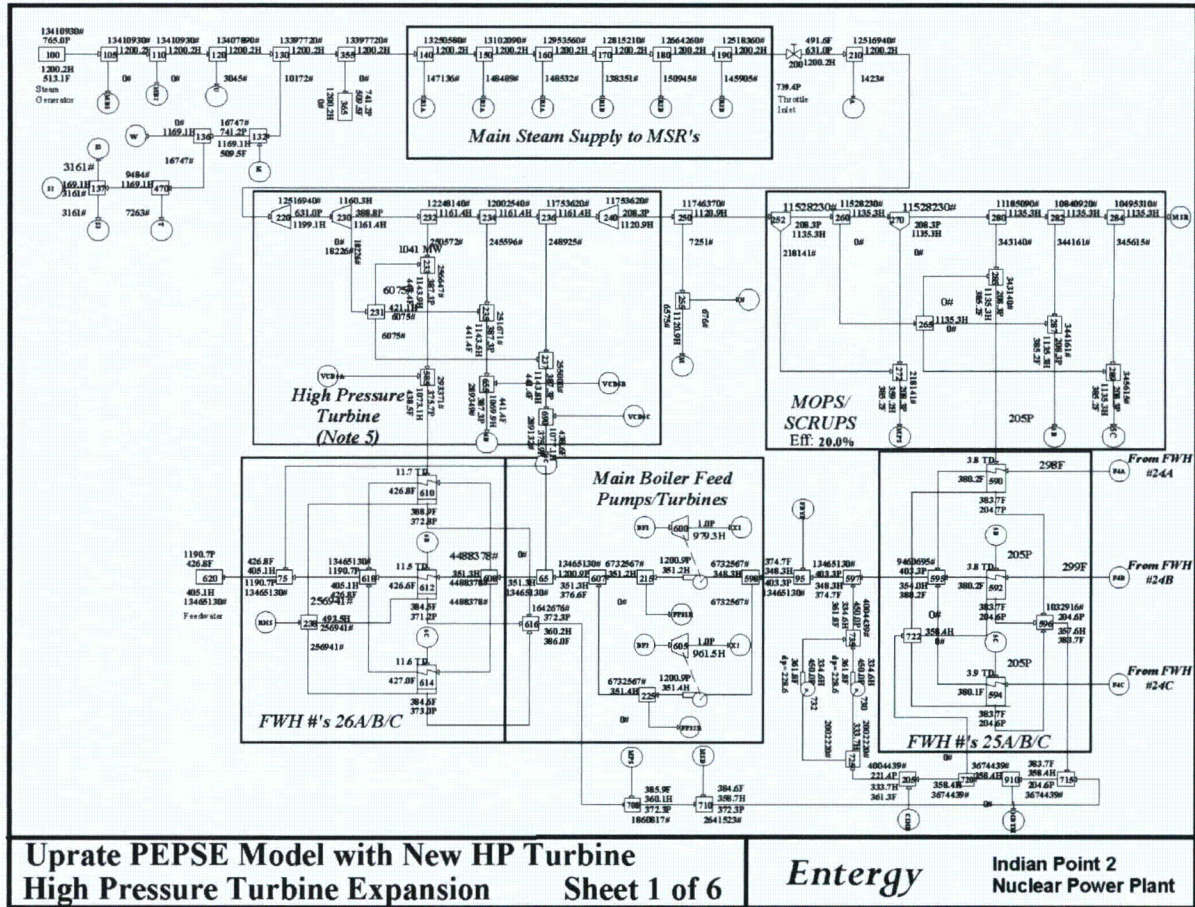


Figure 4-1 IPEC Unit 2 PEPSE Model, Sheet 1

The main components included in Sheet 1 are the input component for the Reactor Flow, the High Pressure Turbine components, the Main Steam Supply components to the Moisture Separator Reheater components, the Feedwater Heater 25 A/B/C and 26 A/B/C components, the Main Boiler Feed Pump/Turbine components, and Moisture Pre-Separator / Special Cross Under Pipe Separator components.

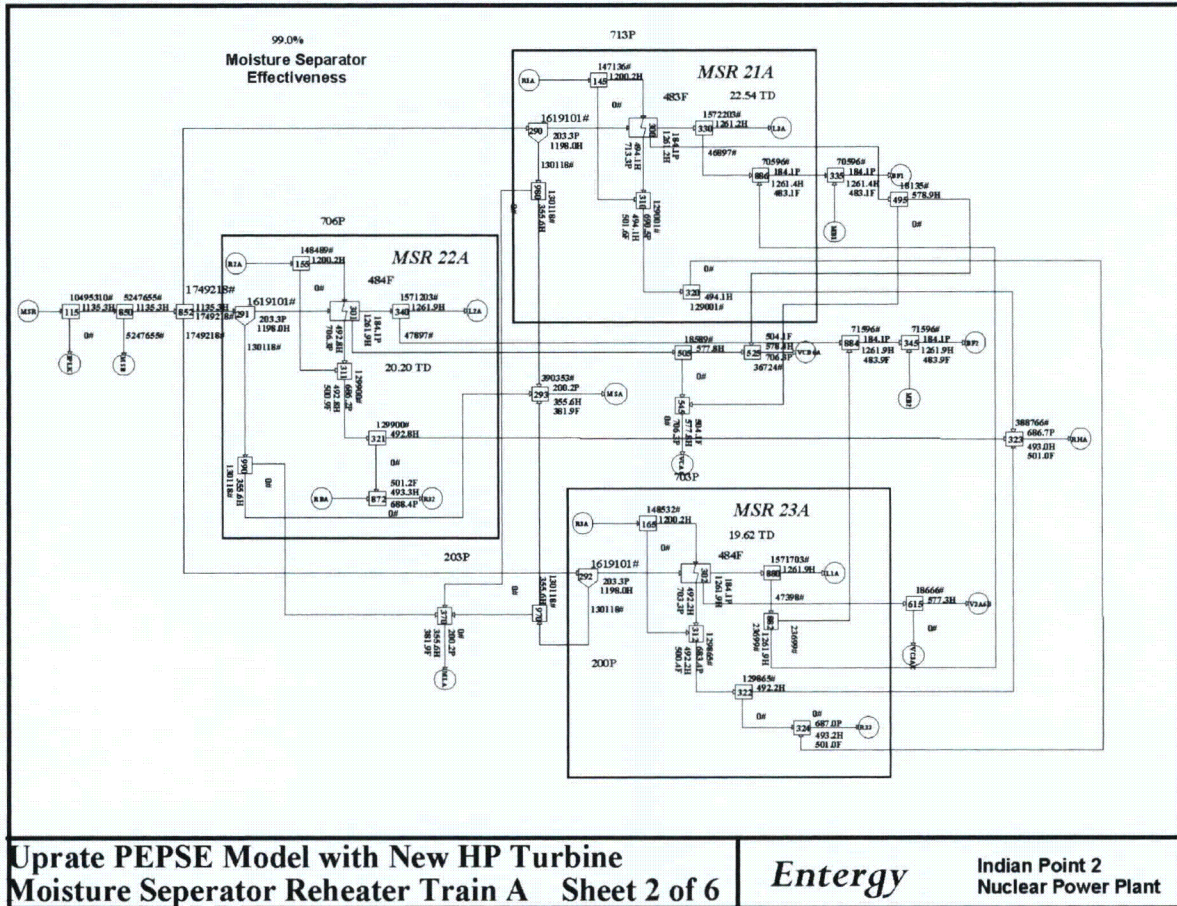


Figure 4-2 IPEC Unit 2 PEPSE Model, Sheet 2

The main components included in Sheet 2 are the Moisture Separator Reheater 21A, 22A, and 23A components.

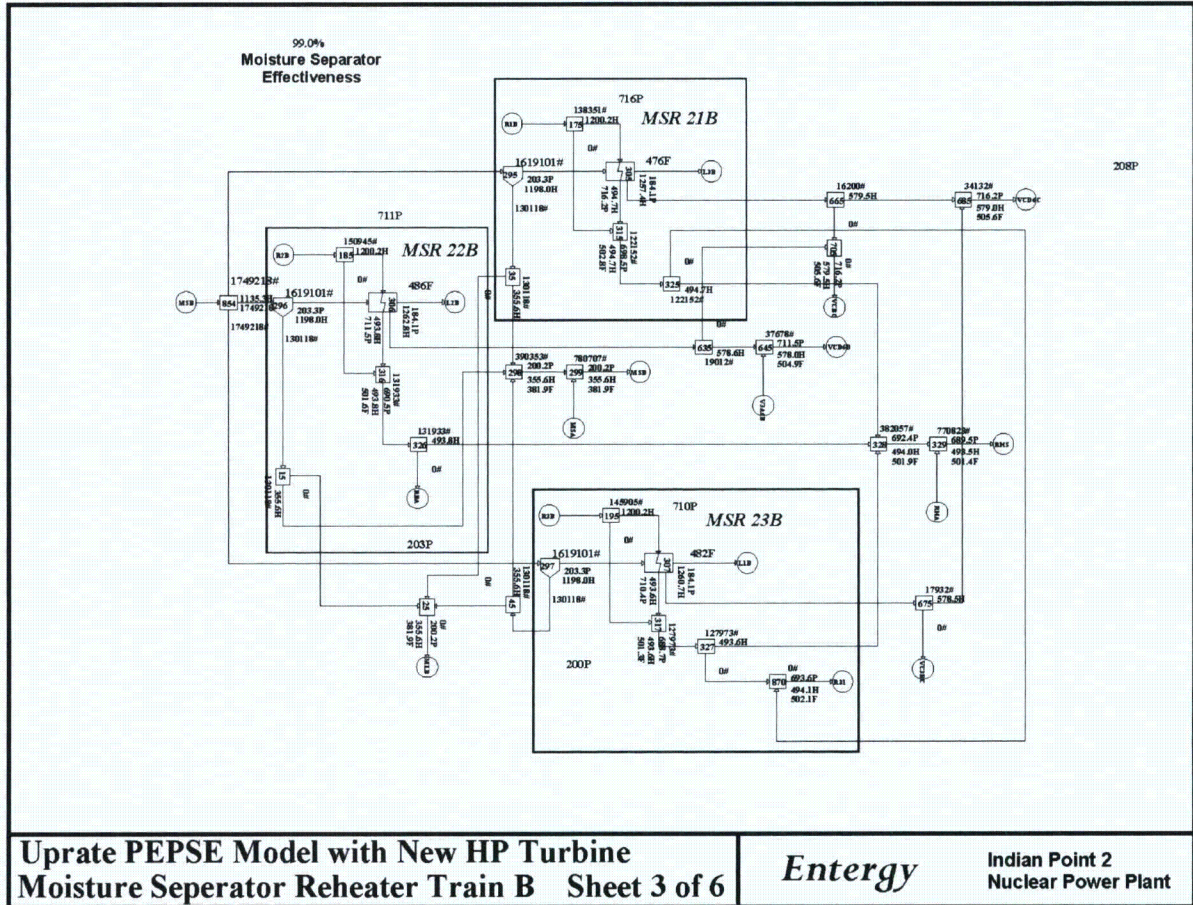
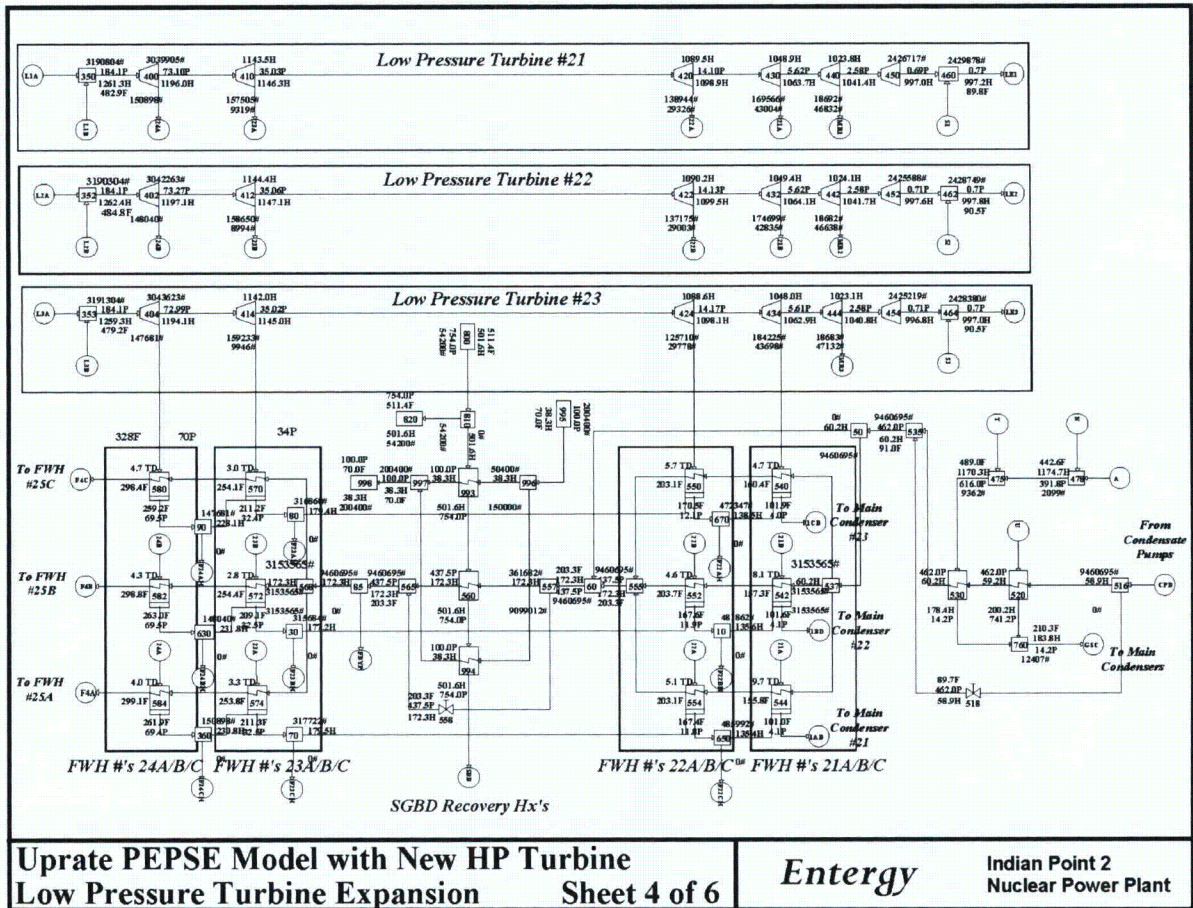


Figure 4-3 IPEC Unit 2 PEPSE Model, Sheet 3

The main components included in Sheet 3 are the Moisture Separator Reheater 21B, 22B, and 23B components.



Uprate PEPSE Model with New HP Turbine  
Low Pressure Turbine Expansion Sheet 4 of 6

Entergy

Indian Point 2  
Nuclear Power Plant

Figure 4-4 IPEC Unit 2 PEPSE Model, Sheet 4

The main components included in Sheet 4 are the Low Pressure Turbine 21, 22, and 23 components, and the Feedwater Heater 21 A/B/C, 22 A/B/C, 23 A/B/C, and 24 A/B/C components.

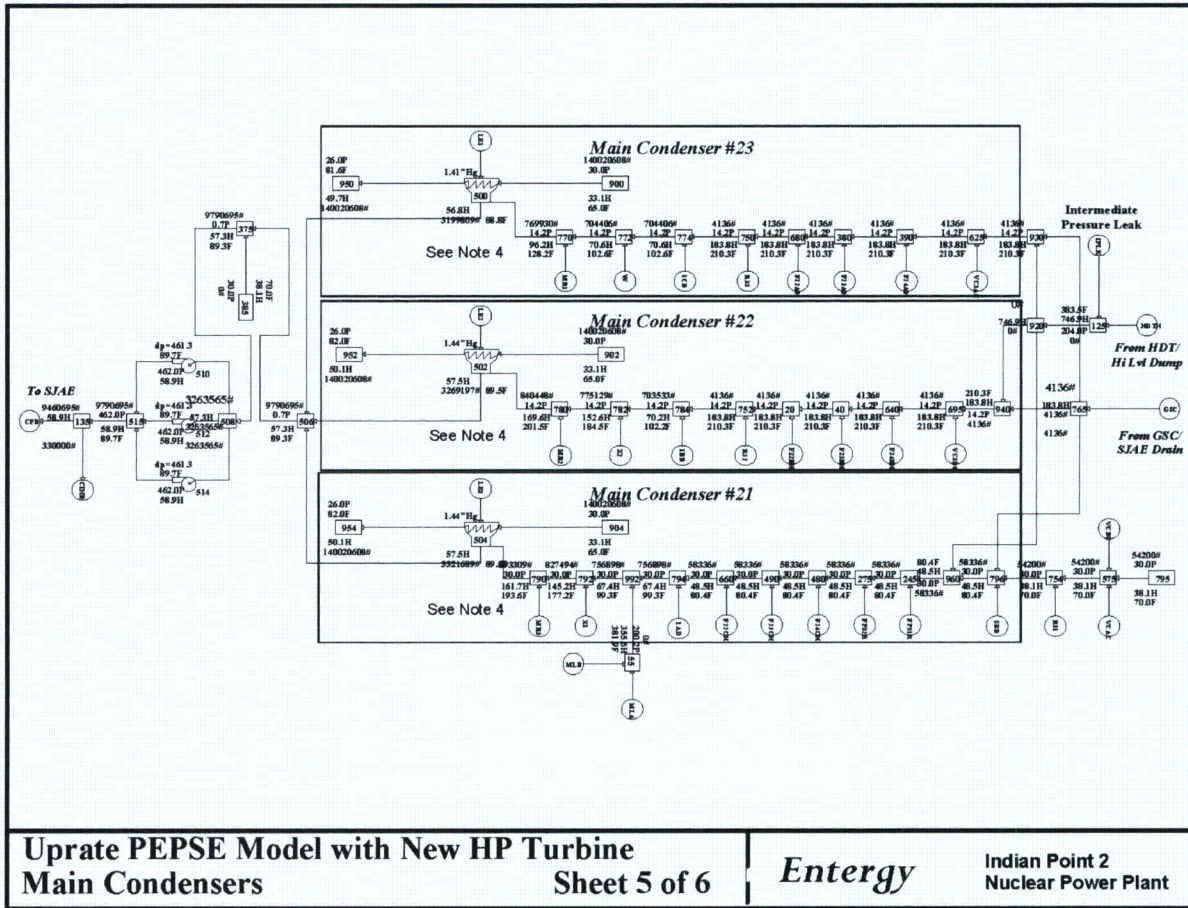


Figure 4-5 IPEC Unit 2 PEPSE Model, Sheet 5

The main components included in Sheet 5 are the Main Condenser 21, 22, and 23 components.

<p>NOTES:</p> <ol style="list-style-type: none"> <li>1. This model runs Data Sets 1, 6, 5 and 16 in that order. S&amp;W changes are in Data Set 16, or else in the last set where the data are entered.</li> <li>2. OPVB 12 is the input for Circ Water Inlet Temp. Operations 103-105 set this value for the CW sources.</li> <li>3. This heat balance should not be used to predict pressures in the condensate and feedwater system. The hydraulic performance has not been tuned to reflect actual plant conditions.</li> <li>4. Hotwell subcooling model updated to use feature of PEPSE 69. Fictitious heat exchangers (Component Numbers 395, 405, 455) deleted from the model. (RMM 4/13/06)</li> <li>5. New HP turbine is tuned to match turbine parameters per Siemens-Westinghouse heat balance WB-9341</li> <li>6. The reheaters model was upgraded to the four pass modeling allowed by of PEPSE 69. (RMM 4/5/05)</li> </ol>	
<p><b>Uprate PEPSE Model with New HP Turbine Notes</b></p>	<p style="text-align: right;"><b>Sheet 6 of 6</b></p>
<p><i>Entergy</i>      <b>Indian Point 2 Nuclear Power Plant</b></p>	

Figure 4-6                      IPEC Unit 2 PEPSE Model, Sheet 6

Sheet 6 includes the notes associated with the Unit 3 PEPSE Model. Notes detail the model updates, the data used to benchmark model performance, and the limitations on the model. The notes provided indicate that information gathered from the main steam system is valid for use in this analysis.

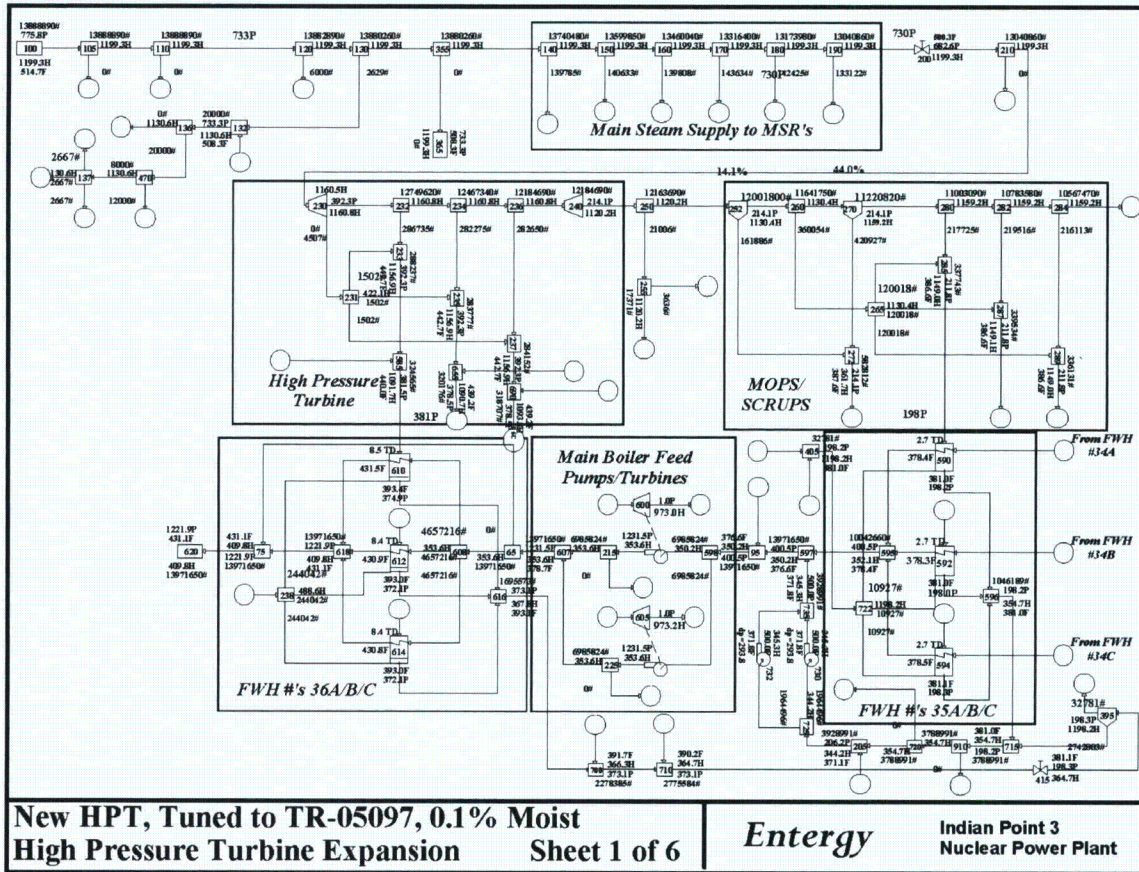
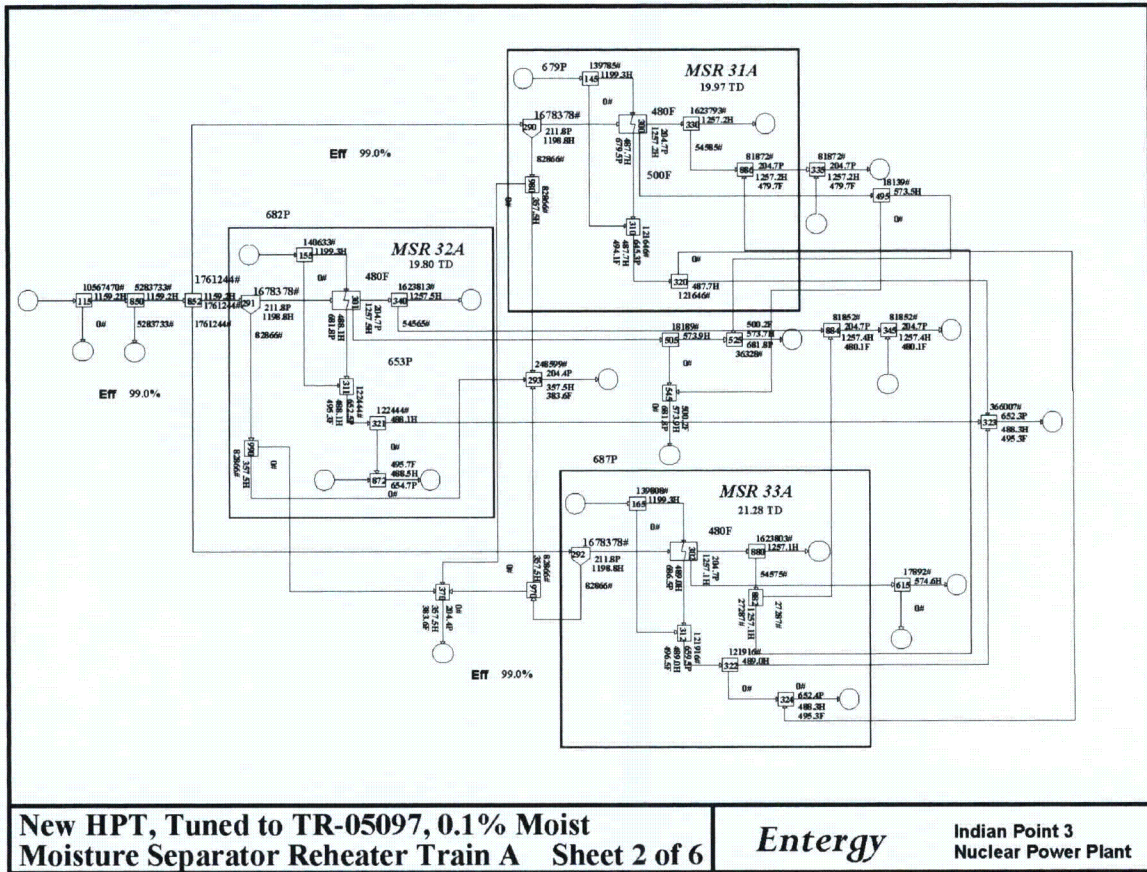


Figure 4-7 IPEC Unit 3 PEPSE Model, Sheet 1

The main components included in Sheet 1 are the input component for the Reactor Flow, the High Pressure Turbine components, the Main Steam Supply components to the Moisture Separator Reheater components, the Feedwater Heater 35 A/B/C and 36 A/B/C components, the Main Boiler Feed Pump/Turbine components, and Moisture Pre-Separator / Special Cross Under Pipe Separator components.



New HPT, Tuned to TR-05097, 0.1% Moist  
Moisture Separator Reheater Train A Sheet 2 of 6

*Energy*

Indian Point 3  
Nuclear Power Plant

Figure 4-8 IPEC Unit 3 PEPSE Model, Sheet 2

The main components included in Sheet 2 are the Moisture Separator Reheater 31A, 32A, and 33A components.



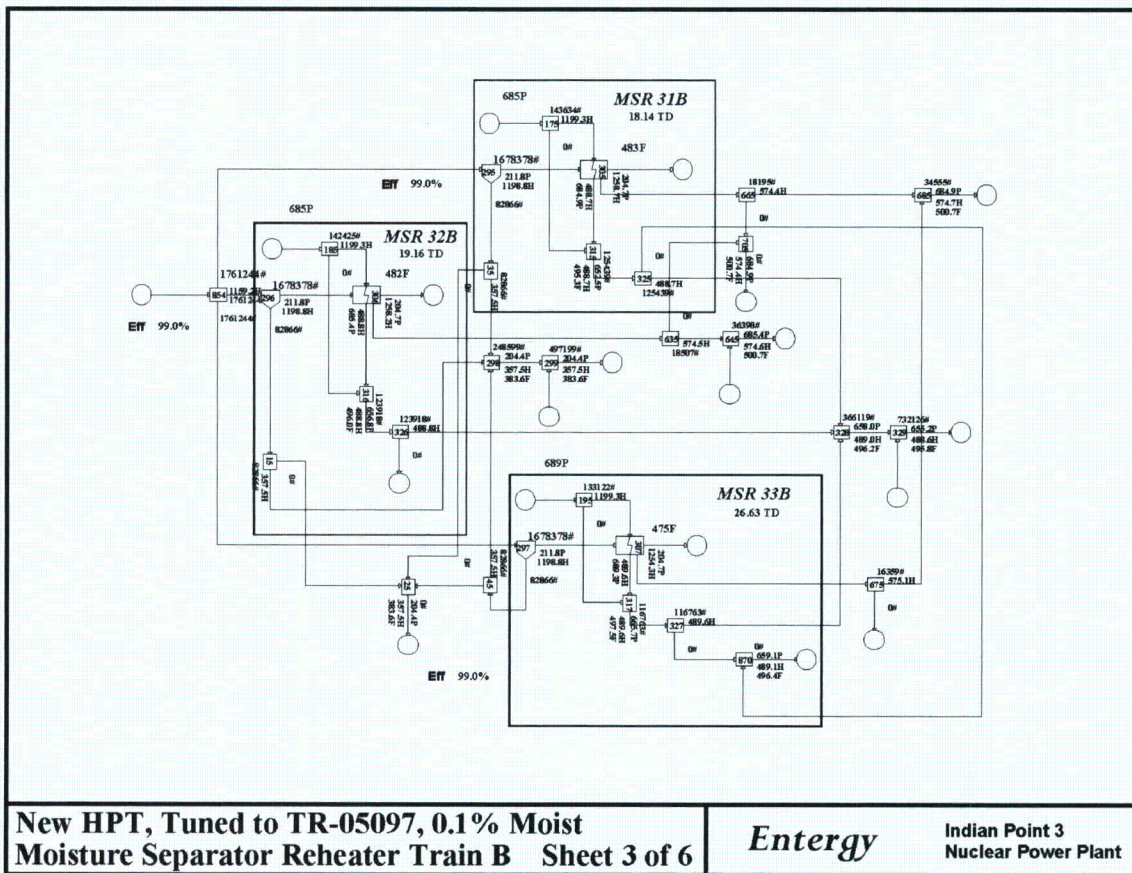


Figure 4-9

IPEC Unit 3 PEPSE Model, Sheet 3

The main components included in Sheet 3 are the Moisture Separator Reheater 31B, 32B, and 33B components.

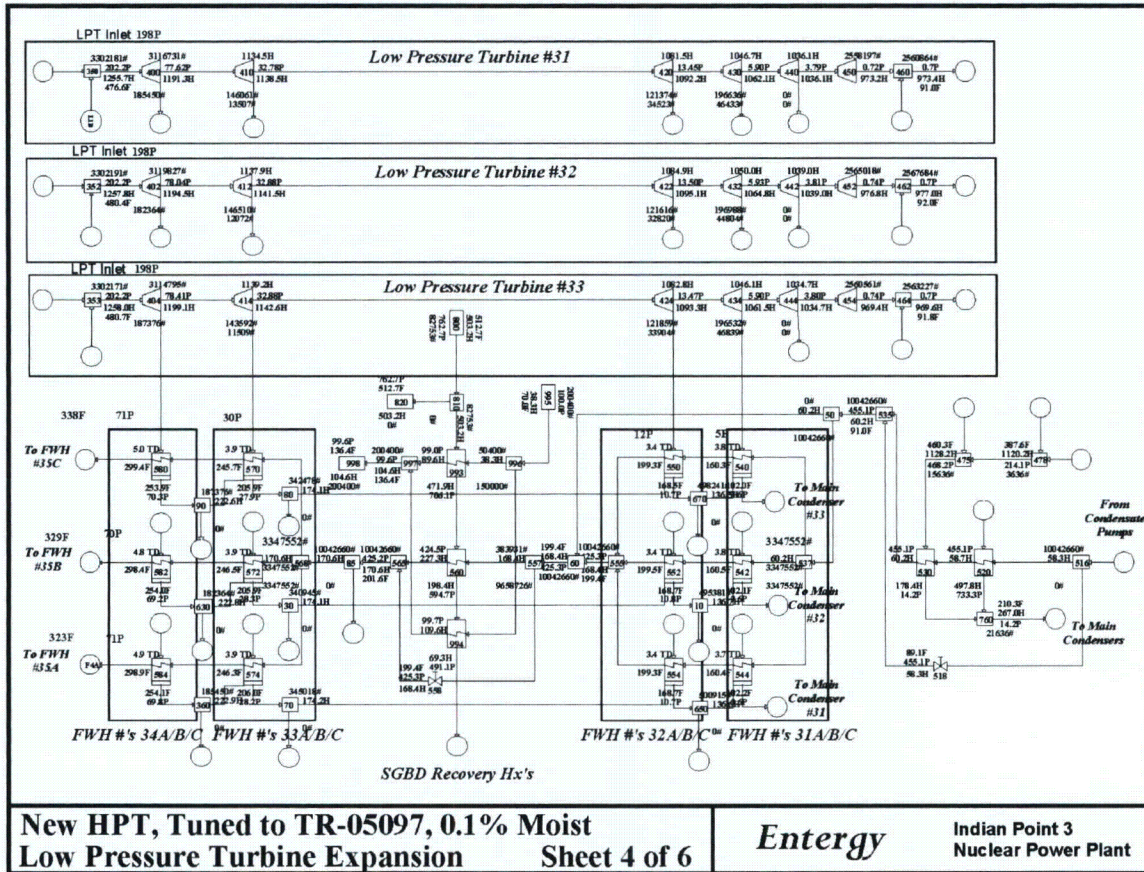


Figure 4-10 IPEC Unit 3 PEPSE Model, Sheet 4

The main components included in Sheet 4 are the Low Pressure Turbine 31, 32, and 33 components, and the Feedwater Heater 31 A/B/C, 32 A/B/C, 33 A/B/C, and 34 A/B/C components.

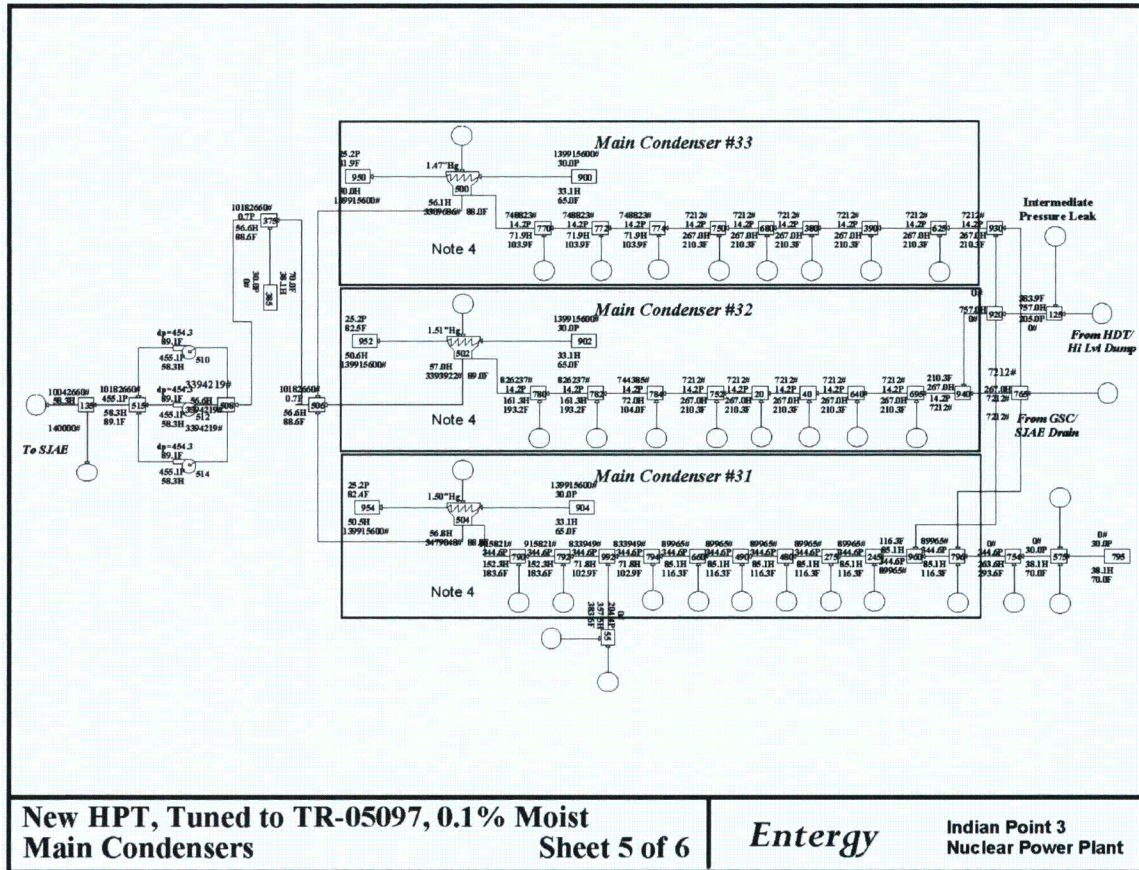


Figure 4-11 IPEC Unit 3 PEPSE Model, Sheet 5

The main components included in Sheet 5 are the Main Condenser 31, 32, and 33 components.

<p>NOTES:</p> <ol style="list-style-type: none"> <li>1. This model runs Data Sets 1, 2, 4, and 11, in that order. S&amp;W changes are in the last data set where the respective data are contained.</li> <li>2. OPVB 12 is the input for Circ Water Inlet Temperature. Operations 103 - 105 apply this value to the three CW source components.</li> <li>3. This heat balance should not be used to predict pressures in the condensate and feedwater systems. The hydraulic performance has not been tuned to reflect actual plant conditions.</li> <li>4. Hotwell subcooling model updated to use feature of PEPSE 69. Fictitious heat exchangers (Component Numbers 425, 435, 445) deleted from the model. (RMM 4/13/06)</li> <li>5. The reheaters modele was upgraded to the four pass modeling allowed by of PEPSE 69. (RMM 8/31/6)</li> <li>6. HP Turbine modeled without dummy Governing Stage. (RMM 9/21/06)</li> <li>7. HP Turbine inlet Bowl Coefficient changed to match projected new Turbine Inlet pressure of 678.9 Psia. (RMM 1/18/07)</li> </ol>	<p style="text-align: center;"><b>Entergy</b></p> <p style="text-align: center;">Indian Point 3 Nuclear Power Plant</p>
<p><b>New HPT, Tuned to TR-05097, 0.1% Moist Notes</b></p>	<p style="text-align: center;"><b>Sheet 6 of 6</b></p>

Figure 4-12                      IPEC Unit 3 PEPSE Model, Sheet 6

Sheet 6 includes the notes associated with the Unit 3 PEPSE Model. Notes detail the model updates, the data used to benchmark model performance, and the limitations on the model. The notes provided indicate that information gathered from the main steam system is valid for use in this analysis.

**Indian Point Unit 2 Average Closed-Loop Cooling Operational Losses (MWe)**

Month	2001	2002	2003	2004	2005	2006	2007	2008
January	4.7	7.6	3.1	2.9	4.2	8.1	8.2	6.6
February	5.5	7.3	3.7	5.1	4.0	6.3	3.3	6.6
March	7.3	9.8	5.6	10.1	5.2	9.6	9.1	9.2
April	15.9	18.4	9.4	16.9	15.3	17.4	15.1	18.3
May	23.7	22.3	11.4	23.5	18.4	23.4	23.4	20.2
June	25.7	24.9	8.2	13.1	19.5	24.4	18.1	20.4
July	11.6	13.3	0.4	8.5	6.3	20.6	9.3	8.1
August	11.3	6.8	0.0	11.6	3.5	7.7	0.0	3.9
September	8.6	10.5	0.0	12.0	2.7	11.6	0.6	7.5
October	13.4	10.7	9.5	14.1	14.2	14.5	13.8	8.6
November	14.9	13.2	15.3	13.3	15.0	18.5	11.2	11.5
December	10.2	6.4	7.2	7.7	6.1	11.7	7.4	6.3
Annual	12.8	12.6	6.2	11.6	9.6	14.5	11.4	10.6

**Indian Point Unit 3 Average Closed-Loop Cooling Operational Losses (MWe)**

Month	2001	2002	2003	2004	2005	2006	2007	2008
January	0.0	0.1	0.0	0.0	0.0	0.2	0.6	0.1
February	0.0	0.1	0.0	0.0	0.0	0.1	0.0	0.2
March	0.0	0.4	0.0	0.3	0.0	0.5	0.5	0.2
April	2.2	5.2	0.6	3.7	3.2	4.1	2.0	4.4
May	10.7	9.6	3.1	13.2	7.4	10.9	11.5	8.7
June	17.8	16.2	3.9	10.3	14.8	17.0	13.8	15.1
July	9.8	11.7	0.3	7.5	5.7	17.4	8.2	7.3
August	10.3	6.3	0.0	10.2	3.4	7.0	0.0	3.4
September	6.9	8.5	0.0	9.3	2.5	8.6	0.5	6.2
October	7.2	5.6	3.9	6.5	7.1	7.1	8.8	4.2
November	4.7	2.4	3.6	3.2	3.4	4.8	2.7	2.7
December	1.9	0.0	0.2	0.2	0.0	0.8	0.1	0.2
Annual	6.0	5.5	1.3	5.4	4.0	6.6	4.5	4.4

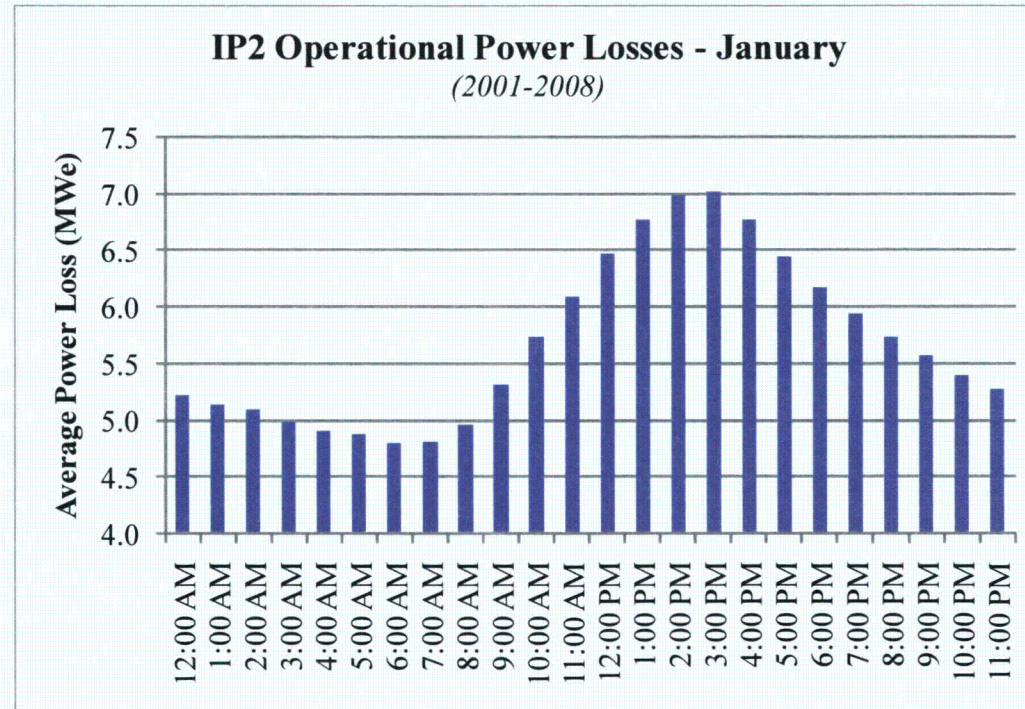
**Indian Point Unit 2 Maximum Closed-Loop Cooling Operational Losses (MWe)**

Month	2001	2002	2003	2004	2005	2006	2007	2008
January	13.9	24.7	12.2	18.4	17.2	29.4	36.7	25.8
February	20.3	20.8	12.3	13.8	11.5	26.2	11.3	28.9
March	13.1	25.3	17.4	29.4	14.6	29.7	31.8	27.5
April	39.6	42.4	29.5	36.1	33.8	35.0	33.2	43.3
May	38.8	43.6	24.8	43.2	33.9	42.9	41.9	42.2
June	39.5	45.8	22.3	33.6	36.6	42.9	38.1	47.4
July	28.1	34.8	9.8	24.9	21.6	34.5	27.2	19.0
August	24.1	21.5	0.0	26.8	17.9	27.5	0.0	17.2
September	25.2	25.1	3.7	27.4	14.3	28.3	3.9	27.1
October	33.5	29.9	31.6	29.5	28.1	34.1	30.8	25.7
November	30.9	37.6	38.3	31.0	34.0	39.5	30.0	28.5
December	27.3	21.7	24.5	26.4	17.1	39.1	27.4	24.4
Annual	39.6	45.8	38.3	43.2	36.6	42.9	41.9	47.4

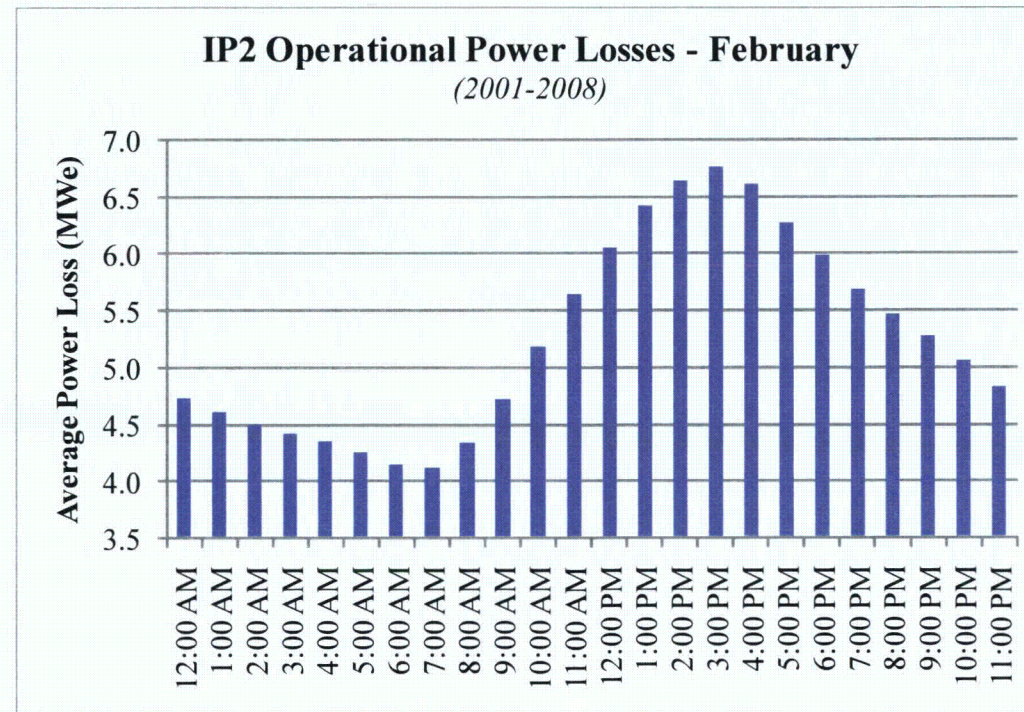
**Indian Point Unit 3 Maximum Closed-Loop Cooling Operational Losses (MWe)**

Month	2001	2002	2003	2004	2005	2006	2007	2008
January	0.0	6.2	0.0	1.7	1.0	9.7	16.1	6.8
February	2.9	3.4	0.0	0.0	0.0	7.2	0.0	9.2
March	0.0	6.6	1.2	9.8	0.0	10.1	11.7	8.2
April	19.3	24.1	9.9	17.6	16.9	16.2	13.0	22.1
May	22.8	29.2	10.6	29.3	19.2	30.4	28.1	28.3
June	31.5	33.0	14.8	29.6	29.4	32.1	31.1	38.0
July	26.6	32.2	7.8	23.6	21.7	32.2	26.0	17.4
August	23.9	21.0	0.0	25.5	17.8	27.6	0.0	15.3
September	22.2	21.5	2.6	22.8	13.6	22.3	2.9	25.2
October	21.8	24.9	17.4	18.6	21.3	20.8	23.7	15.2
November	16.4	20.2	21.2	14.0	16.2	20.4	14.0	13.0
December	12.0	4.0	6.1	7.4	1.0	20.1	8.2	6.1
Annual	31.5	33.0	21.2	29.6	29.4	32.2	31.1	38.0

Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	5.2	29.3
1:00 AM	5.1	29.8
2:00 AM	5.1	30.3
3:00 AM	5.0	30.7
4:00 AM	4.9	31.0
5:00 AM	4.9	30.7
6:00 AM	4.8	34.0
7:00 AM	4.8	33.8
8:00 AM	5.0	35.0
9:00 AM	5.3	36.3
10:00 AM	5.7	36.7
11:00 AM	6.1	34.8
12:00 PM	6.5	33.5
1:00 PM	6.8	32.8
2:00 PM	7.0	31.2
3:00 PM	7.0	29.0
4:00 PM	6.8	27.1
5:00 PM	6.4	27.3
6:00 PM	6.2	26.6
7:00 PM	5.9	28.7
8:00 PM	5.7	29.0
9:00 PM	5.6	29.1
10:00 PM	5.4	28.0
11:00 PM	5.3	29.2

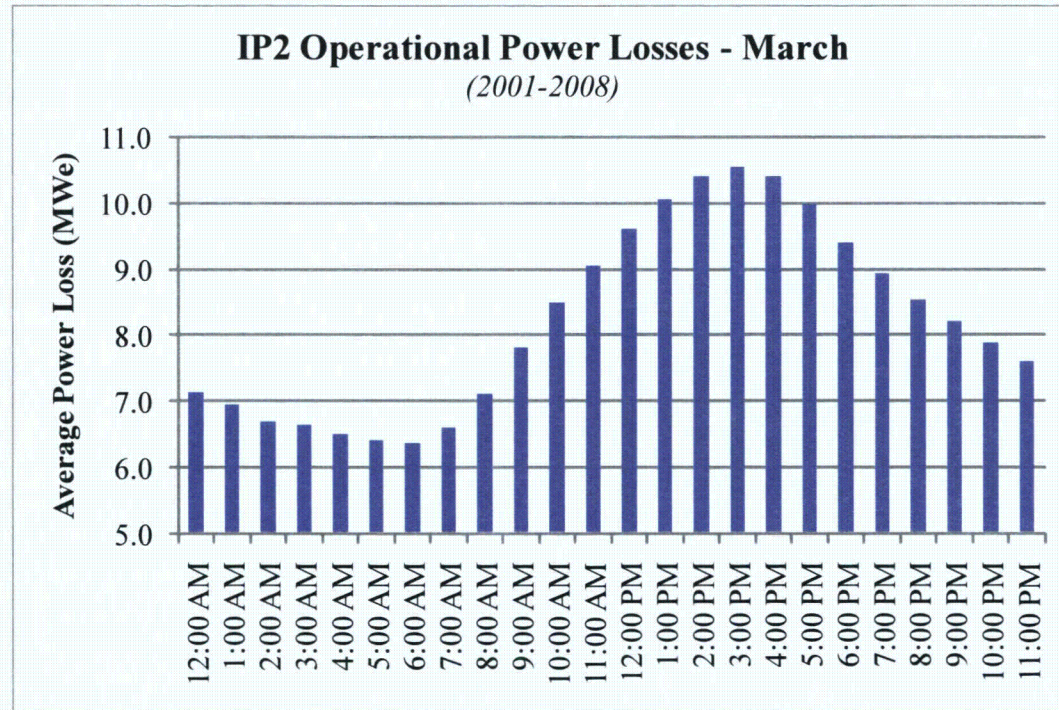


Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	4.7	19.5
1:00 AM	4.6	20.1
2:00 AM	4.5	21.5
3:00 AM	4.4	23.9
4:00 AM	4.4	23.9
5:00 AM	4.3	24.0
6:00 AM	4.1	24.2
7:00 AM	4.1	24.8
8:00 AM	4.3	27.3
9:00 AM	4.7	27.5
10:00 AM	5.2	28.3
11:00 AM	5.6	28.7
12:00 PM	6.1	28.2
1:00 PM	6.4	28.9
2:00 PM	6.6	24.0
3:00 PM	6.8	22.7
4:00 PM	6.6	21.4
5:00 PM	6.3	19.2
6:00 PM	6.0	18.0
7:00 PM	5.7	17.3
8:00 PM	5.5	16.8
9:00 PM	5.3	19.7
10:00 PM	5.1	18.1
11:00 PM	4.8	20.1

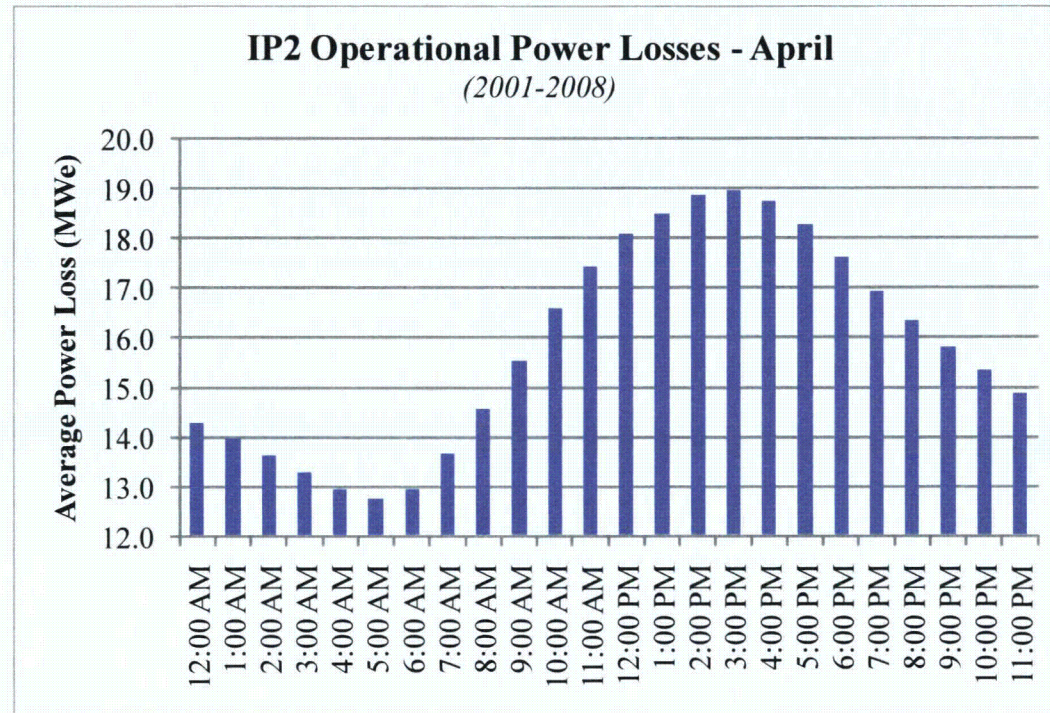




Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	7.1	25.4
1:00 AM	7.0	25.3
2:00 AM	6.7	22.7
3:00 AM	6.6	23.1
4:00 AM	6.5	27.5
5:00 AM	6.4	26.5
6:00 AM	6.4	24.1
7:00 AM	6.6	22.6
8:00 AM	7.1	21.3
9:00 AM	7.8	26.6
10:00 AM	8.5	31.2
11:00 AM	9.0	31.0
12:00 PM	9.6	31.4
1:00 PM	10.0	31.3
2:00 PM	10.4	31.5
3:00 PM	10.5	31.8
4:00 PM	10.4	31.5
5:00 PM	10.0	29.5
6:00 PM	9.4	27.7
7:00 PM	8.9	25.3
8:00 PM	8.5	24.8
9:00 PM	8.2	25.6
10:00 PM	7.9	26.1
11:00 PM	7.6	25.6



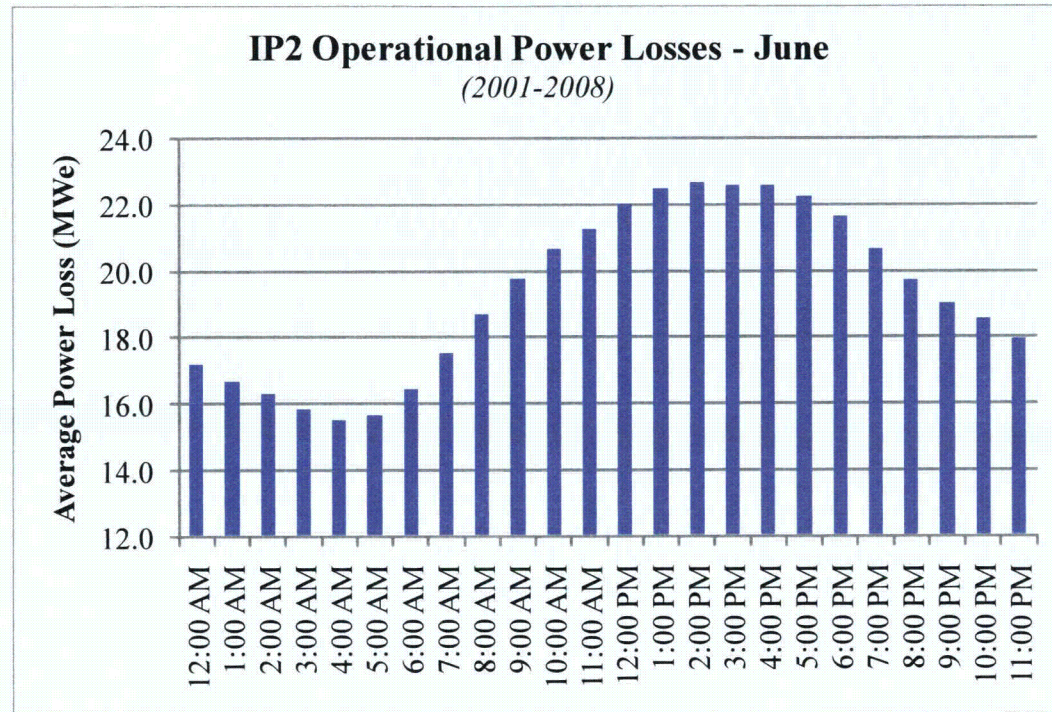
Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	14.3	37.4
1:00 AM	14.0	36.9
2:00 AM	13.6	35.7
3:00 AM	13.3	35.1
4:00 AM	12.9	35.1
5:00 AM	12.8	34.6
6:00 AM	12.9	35.6
7:00 AM	13.7	38.2
8:00 AM	14.6	40.1
9:00 AM	15.5	41.4
10:00 AM	16.6	42.4
11:00 AM	17.4	41.7
12:00 PM	18.1	42.4
1:00 PM	18.5	42.0
2:00 PM	18.9	43.3
3:00 PM	19.0	41.8
4:00 PM	18.7	41.7
5:00 PM	18.3	41.1
6:00 PM	17.6	41.5
7:00 PM	16.9	40.1
8:00 PM	16.3	39.3
9:00 PM	15.8	38.4
10:00 PM	15.3	37.7
11:00 PM	14.9	37.6



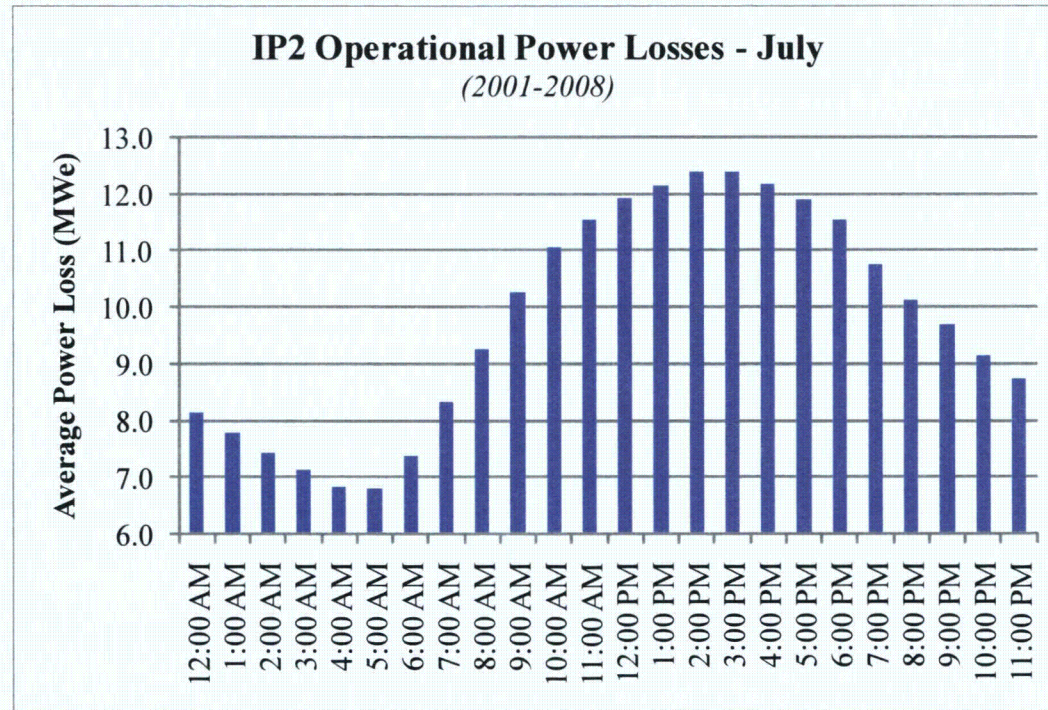
Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	18.7	35.4
1:00 AM	18.4	35.0
2:00 AM	18.0	34.6
3:00 AM	17.7	33.9
4:00 AM	17.4	32.6
5:00 AM	17.4	33.0
6:00 AM	18.0	34.1
7:00 AM	18.9	37.3
8:00 AM	20.0	40.0
9:00 AM	21.0	41.1
10:00 AM	22.0	41.8
11:00 AM	22.7	42.7
12:00 PM	23.3	43.1
1:00 PM	23.8	43.6
2:00 PM	24.0	43.6
3:00 PM	24.2	43.3
4:00 PM	24.0	42.9
5:00 PM	23.5	41.8
6:00 PM	22.7	39.7
7:00 PM	21.8	38.2
8:00 PM	21.1	37.6
9:00 PM	20.5	36.9
10:00 PM	20.0	36.0
11:00 PM	19.6	35.5



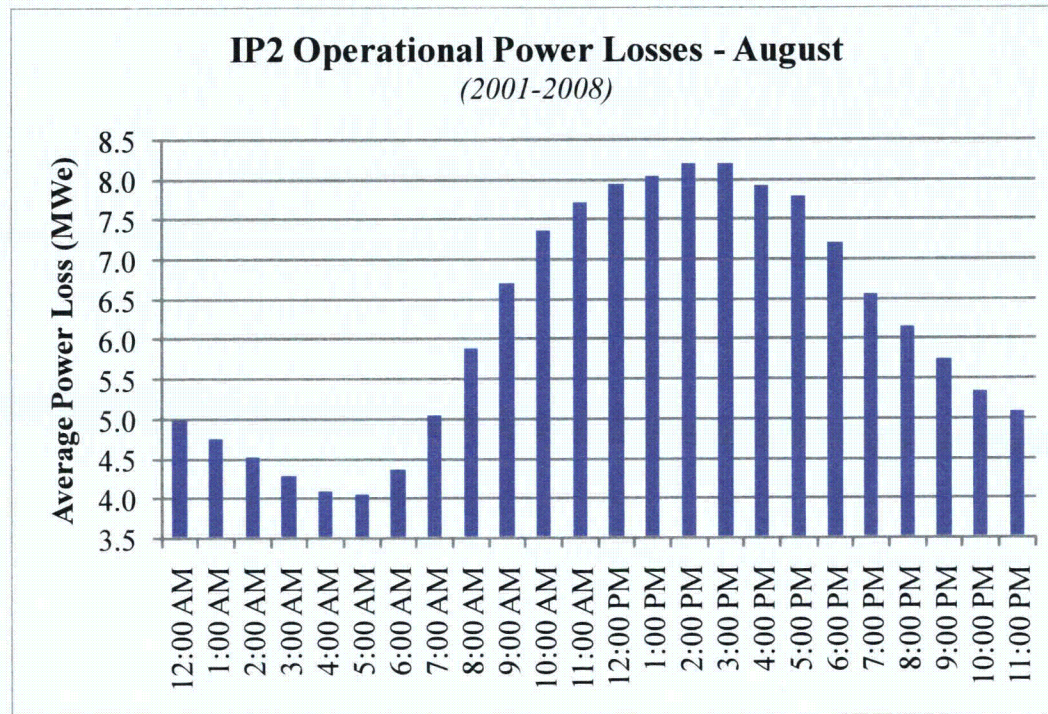
Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	17.2	39.3
1:00 AM	16.7	38.6
2:00 AM	16.3	38.1
3:00 AM	15.8	36.7
4:00 AM	15.5	35.5
5:00 AM	15.7	37.7
6:00 AM	16.4	38.3
7:00 AM	17.5	38.8
8:00 AM	18.7	40.2
9:00 AM	19.8	41.1
10:00 AM	20.7	41.7
11:00 AM	21.3	43.6
12:00 PM	22.0	44.6
1:00 PM	22.5	45.3
2:00 PM	22.7	47.4
3:00 PM	22.6	46.8
4:00 PM	22.6	45.6
5:00 PM	22.2	45.1
6:00 PM	21.6	43.2
7:00 PM	20.6	41.4
8:00 PM	19.7	40.5
9:00 PM	19.0	40.2
10:00 PM	18.5	39.6
11:00 PM	17.9	39.0



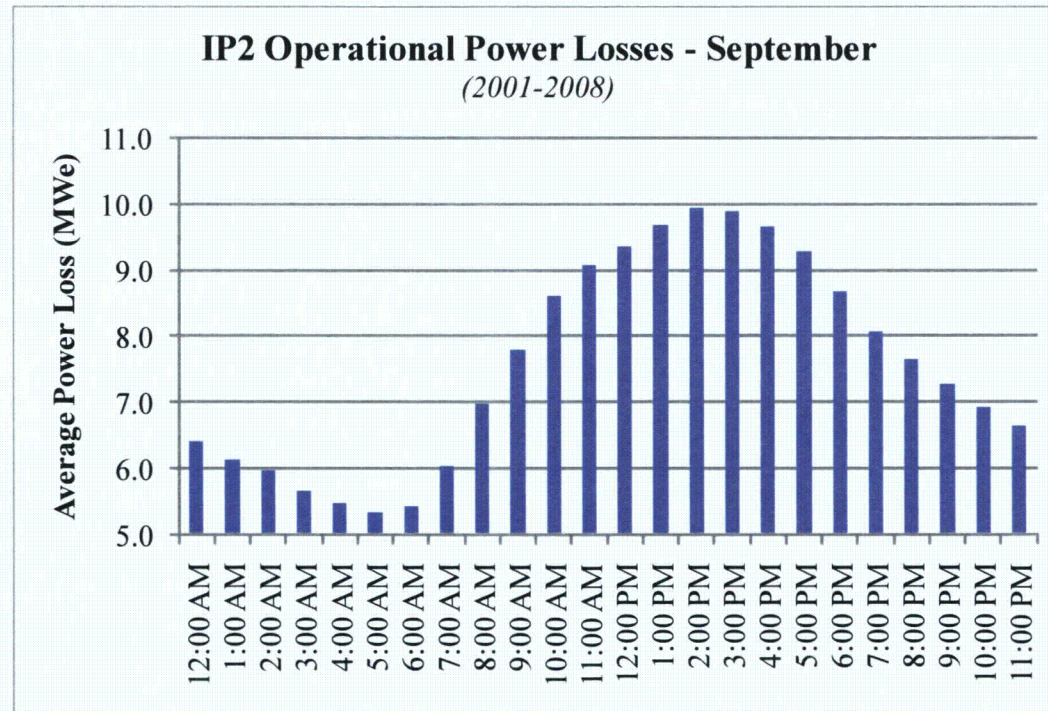
Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	8.1	29.2
1:00 AM	7.8	27.7
2:00 AM	7.4	27.1
3:00 AM	7.1	26.6
4:00 AM	6.8	26.2
5:00 AM	6.8	26.2
6:00 AM	7.4	26.9
7:00 AM	8.3	27.9
8:00 AM	9.3	28.8
9:00 AM	10.3	30.0
10:00 AM	11.0	31.5
11:00 AM	11.5	33.4
12:00 PM	11.9	34.0
1:00 PM	12.2	33.8
2:00 PM	12.4	33.7
3:00 PM	12.4	34.8
4:00 PM	12.2	34.7
5:00 PM	11.9	34.7
6:00 PM	11.5	34.5
7:00 PM	10.7	32.9
8:00 PM	10.1	31.4
9:00 PM	9.7	31.5
10:00 PM	9.1	32.0
11:00 PM	8.7	31.9



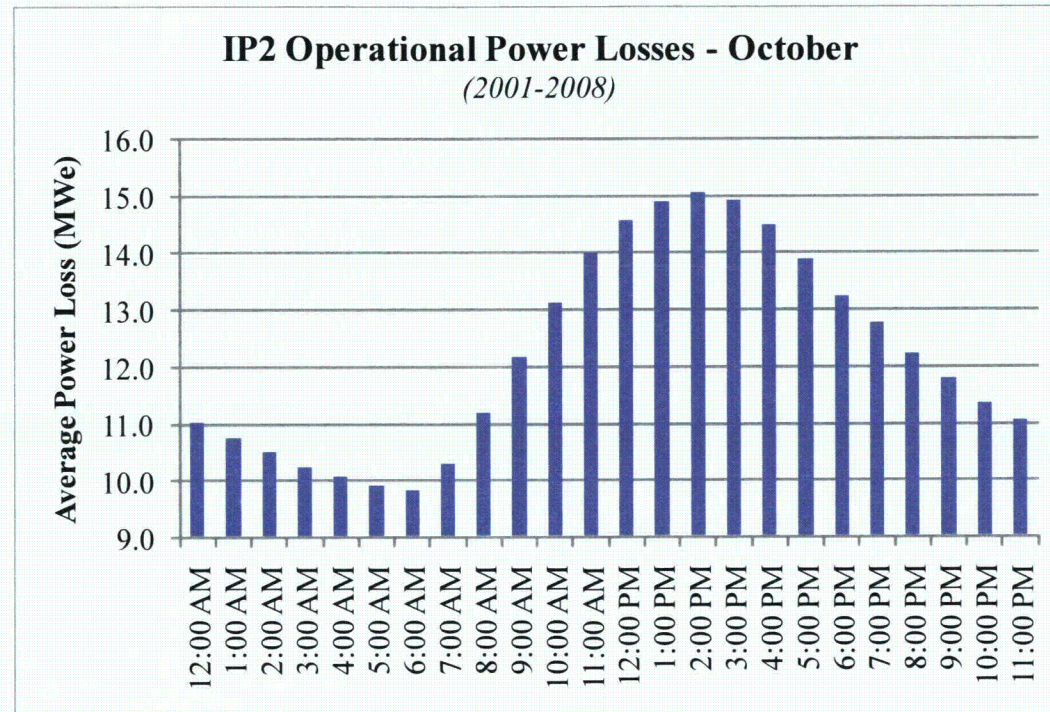
Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	5.0	21.2
1:00 AM	4.8	21.4
2:00 AM	4.5	20.6
3:00 AM	4.3	20.0
4:00 AM	4.1	19.6
5:00 AM	4.0	20.2
6:00 AM	4.4	21.1
7:00 AM	5.0	22.4
8:00 AM	5.9	24.1
9:00 AM	6.7	24.0
10:00 AM	7.4	25.7
11:00 AM	7.7	27.0
12:00 PM	7.9	27.5
1:00 PM	8.0	27.4
2:00 PM	8.2	27.5
3:00 PM	8.2	27.4
4:00 PM	7.9	27.1
5:00 PM	7.8	26.2
6:00 PM	7.2	23.8
7:00 PM	6.6	22.4
8:00 PM	6.2	22.2
9:00 PM	5.7	22.0
10:00 PM	5.3	22.3
11:00 PM	5.1	22.8



Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	6.4	25.1
1:00 AM	6.1	23.3
2:00 AM	6.0	25.2
3:00 AM	5.7	25.2
4:00 AM	5.5	24.1
5:00 AM	5.3	24.6
6:00 AM	5.4	25.6
7:00 AM	6.0	27.2
8:00 AM	7.0	27.5
9:00 AM	7.8	28.3
10:00 AM	8.6	28.2
11:00 AM	9.1	27.5
12:00 PM	9.4	26.6
1:00 PM	9.7	26.7
2:00 PM	9.9	27.1
3:00 PM	9.9	27.4
4:00 PM	9.7	25.9
5:00 PM	9.3	24.4
6:00 PM	8.7	24.4
7:00 PM	8.1	24.7
8:00 PM	7.7	24.7
9:00 PM	7.3	25.0
10:00 PM	6.9	24.6
11:00 PM	6.6	24.0

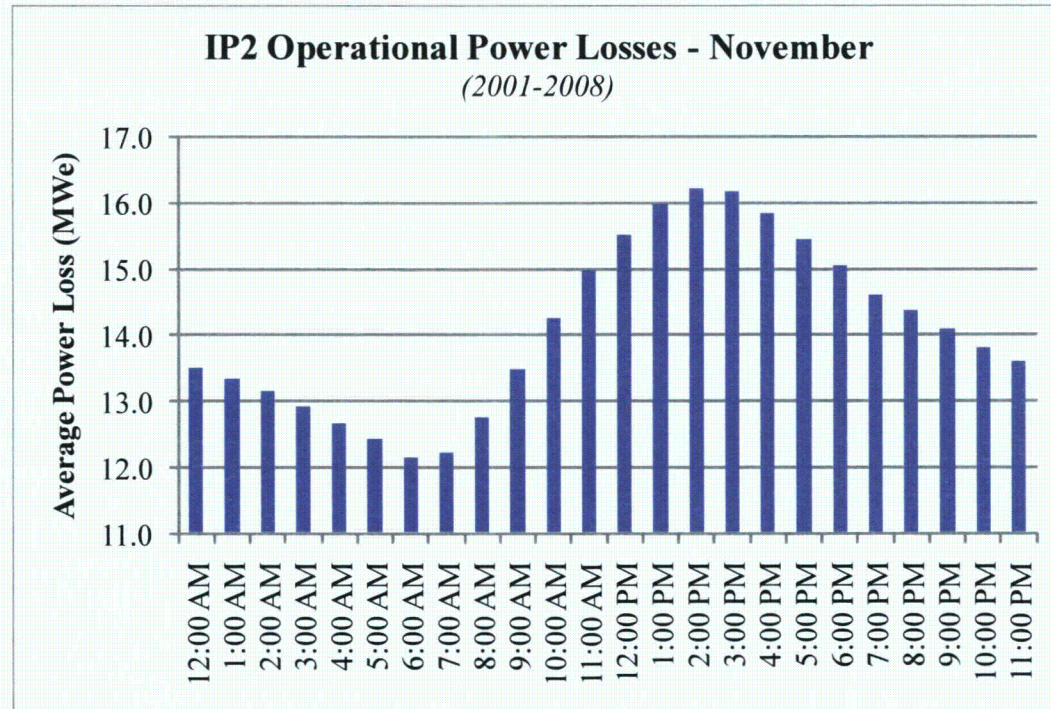


Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	11.0	31.6
1:00 AM	10.7	31.3
2:00 AM	10.5	30.9
3:00 AM	10.2	30.5
4:00 AM	10.1	30.0
5:00 AM	9.9	29.6
6:00 AM	9.8	29.3
7:00 AM	10.3	29.1
8:00 AM	11.2	29.9
9:00 AM	12.2	31.5
10:00 AM	13.1	32.5
11:00 AM	14.0	34.1
12:00 PM	14.6	33.7
1:00 PM	14.9	32.4
2:00 PM	15.1	33.1
3:00 PM	14.9	33.5
4:00 PM	14.5	31.8
5:00 PM	13.9	30.6
6:00 PM	13.2	30.5
7:00 PM	12.8	30.6
8:00 PM	12.2	30.9
9:00 PM	11.8	30.8
10:00 PM	11.4	30.5
11:00 PM	11.1	31.2

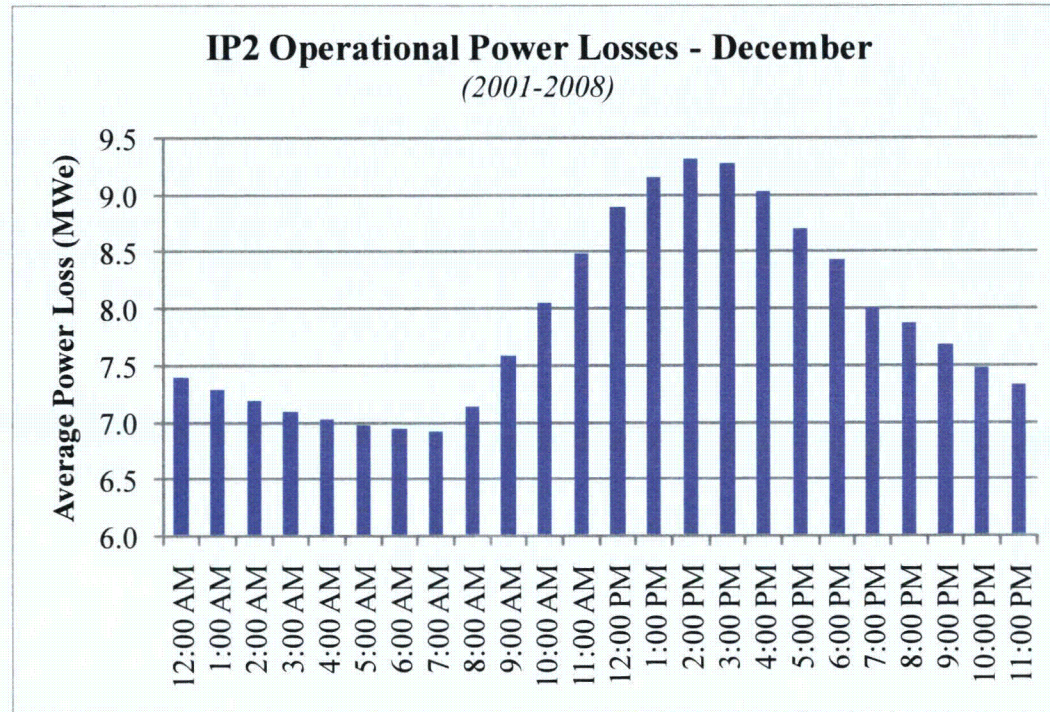




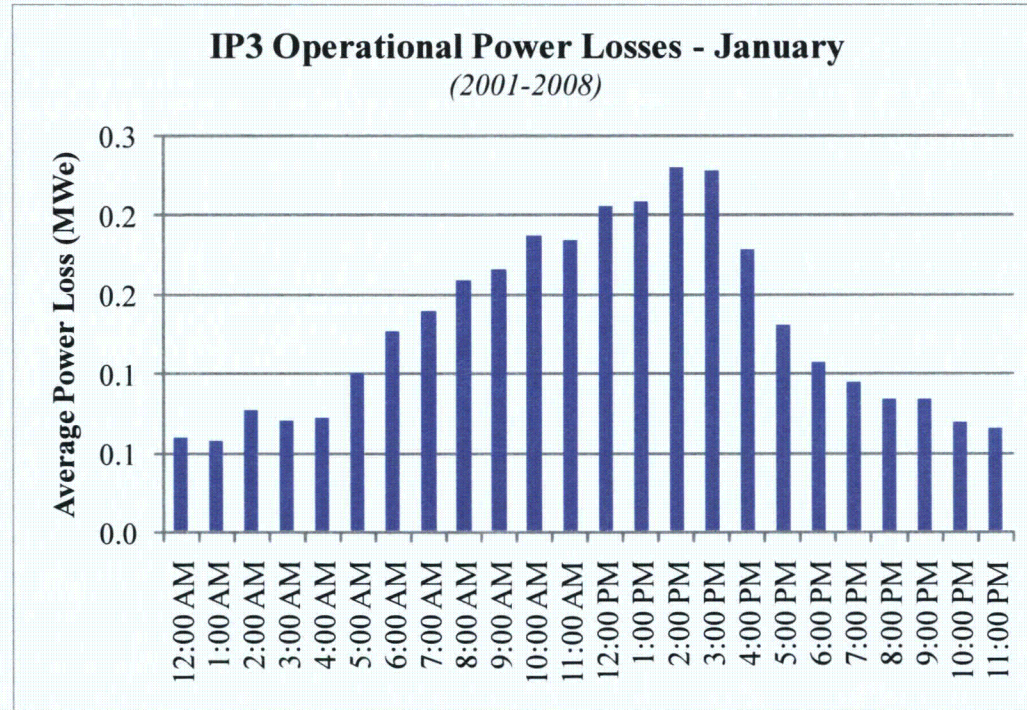
Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	13.5	37.2
1:00 AM	13.3	36.0
2:00 AM	13.2	34.6
3:00 AM	12.9	32.1
4:00 AM	12.7	32.1
5:00 AM	12.4	32.5
6:00 AM	12.2	33.0
7:00 AM	12.2	32.9
8:00 AM	12.8	32.7
9:00 AM	13.5	34.0
10:00 AM	14.3	36.1
11:00 AM	15.0	37.3
12:00 PM	15.5	37.7
1:00 PM	16.0	38.0
2:00 PM	16.2	38.8
3:00 PM	16.2	38.6
4:00 PM	15.8	38.1
5:00 PM	15.5	38.6
6:00 PM	15.1	38.8
7:00 PM	14.6	39.0
8:00 PM	14.4	39.3
9:00 PM	14.1	38.8
10:00 PM	13.8	39.5
11:00 PM	13.6	38.7



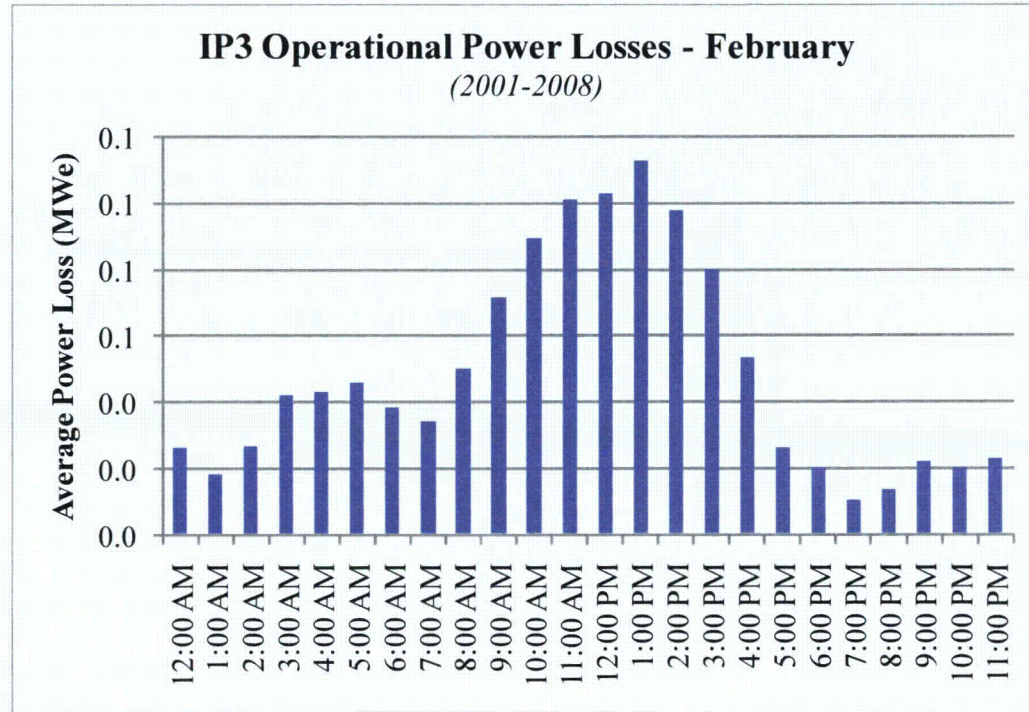
Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	7.4	32.5
1:00 AM	7.3	32.6
2:00 AM	7.2	32.6
3:00 AM	7.1	32.9
4:00 AM	7.0	32.6
5:00 AM	7.0	32.6
6:00 AM	6.9	34.2
7:00 AM	6.9	34.3
8:00 AM	7.1	34.5
9:00 AM	7.6	36.1
10:00 AM	8.0	34.8
11:00 AM	8.5	34.6
12:00 PM	8.9	34.8
1:00 PM	9.1	34.7
2:00 PM	9.3	34.5
3:00 PM	9.3	34.7
4:00 PM	9.0	35.5
5:00 PM	8.7	36.0
6:00 PM	8.4	39.1
7:00 PM	8.0	26.0
8:00 PM	7.9	29.9
9:00 PM	7.7	27.4
10:00 PM	7.5	22.4
11:00 PM	7.3	20.7



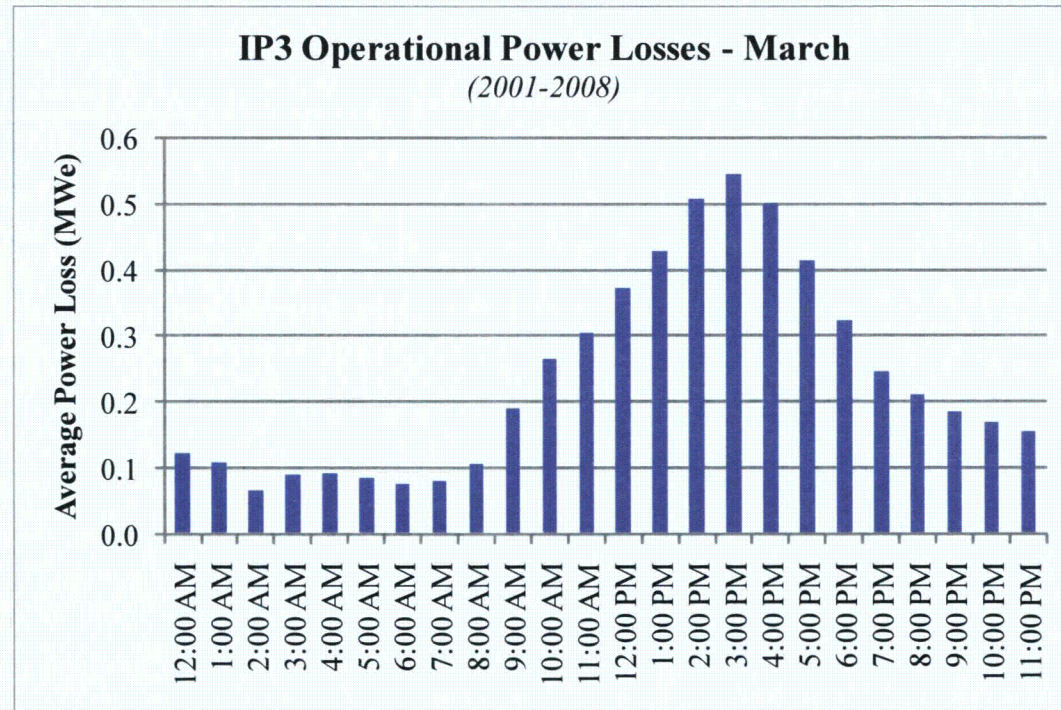
Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	0.1	9.8
1:00 AM	0.1	10.2
2:00 AM	0.1	10.5
3:00 AM	0.1	10.9
4:00 AM	0.1	11.1
5:00 AM	0.1	10.9
6:00 AM	0.1	13.7
7:00 AM	0.1	13.5
8:00 AM	0.2	14.6
9:00 AM	0.2	15.7
10:00 AM	0.2	16.1
11:00 AM	0.2	14.4
12:00 PM	0.2	13.3
1:00 PM	0.2	12.6
2:00 PM	0.2	11.2
3:00 PM	0.2	9.5
4:00 PM	0.2	8.0
5:00 PM	0.1	8.1
6:00 PM	0.1	7.6
7:00 PM	0.1	9.2
8:00 PM	0.1	9.5
9:00 PM	0.1	9.6
10:00 PM	0.1	8.7
11:00 PM	0.1	9.7



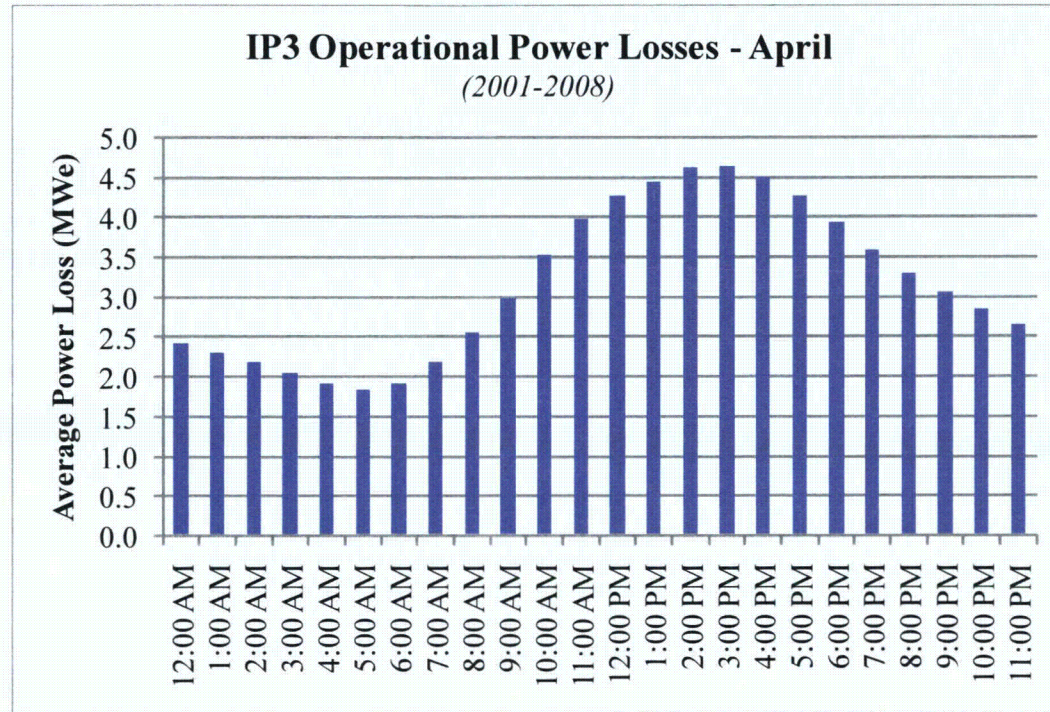
Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	0.0	2.5
1:00 AM	0.0	2.9
2:00 AM	0.0	3.8
3:00 AM	0.0	5.4
4:00 AM	0.0	5.5
5:00 AM	0.0	5.6
6:00 AM	0.0	5.6
7:00 AM	0.0	6.1
8:00 AM	0.1	8.0
9:00 AM	0.1	8.1
10:00 AM	0.1	8.8
11:00 AM	0.1	9.1
12:00 PM	0.1	8.7
1:00 PM	0.1	9.2
2:00 PM	0.1	5.5
3:00 PM	0.1	4.7
4:00 PM	0.1	3.7
5:00 PM	0.0	2.3
6:00 PM	0.0	1.5
7:00 PM	0.0	1.1
8:00 PM	0.0	0.8
9:00 PM	0.0	2.6
10:00 PM	0.0	1.6
11:00 PM	0.0	2.9



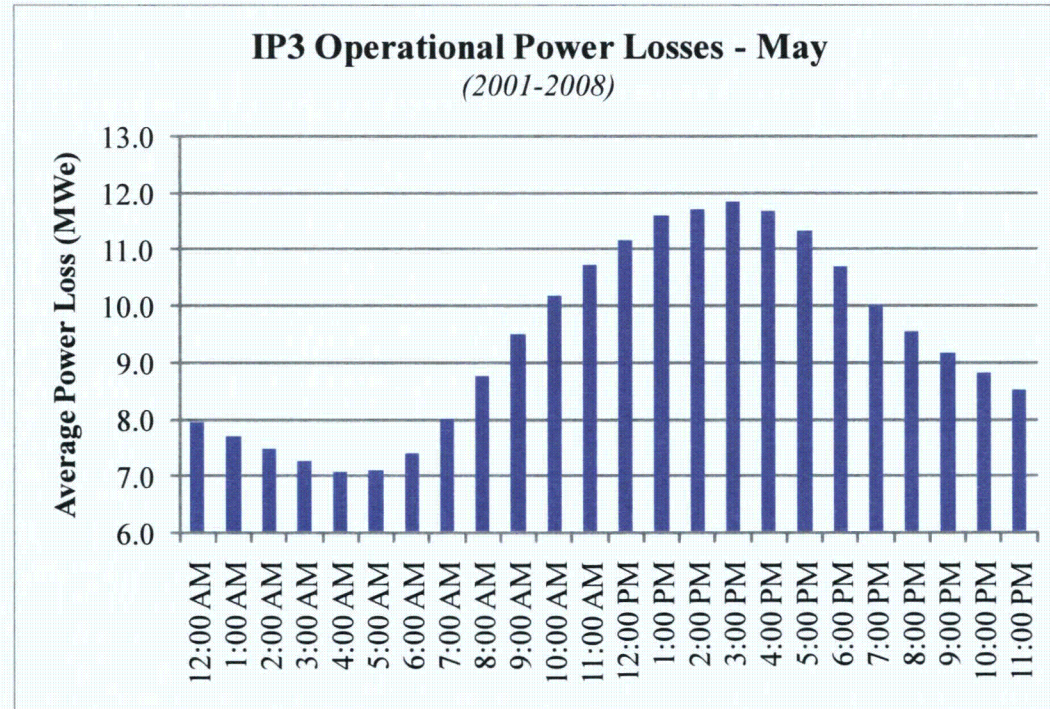
Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	0.1	6.5
1:00 AM	0.1	6.6
2:00 AM	0.1	4.5
3:00 AM	0.1	4.8
4:00 AM	0.1	8.2
5:00 AM	0.1	7.4
6:00 AM	0.1	5.6
7:00 AM	0.1	4.6
8:00 AM	0.1	3.6
9:00 AM	0.2	7.4
10:00 AM	0.3	11.2
11:00 AM	0.3	11.0
12:00 PM	0.4	11.3
1:00 PM	0.4	11.3
2:00 PM	0.5	11.4
3:00 PM	0.5	11.7
4:00 PM	0.5	11.4
5:00 PM	0.4	9.7
6:00 PM	0.3	8.3
7:00 PM	0.2	6.5
8:00 PM	0.2	6.1
9:00 PM	0.2	6.7
10:00 PM	0.2	7.0
11:00 PM	0.2	6.6



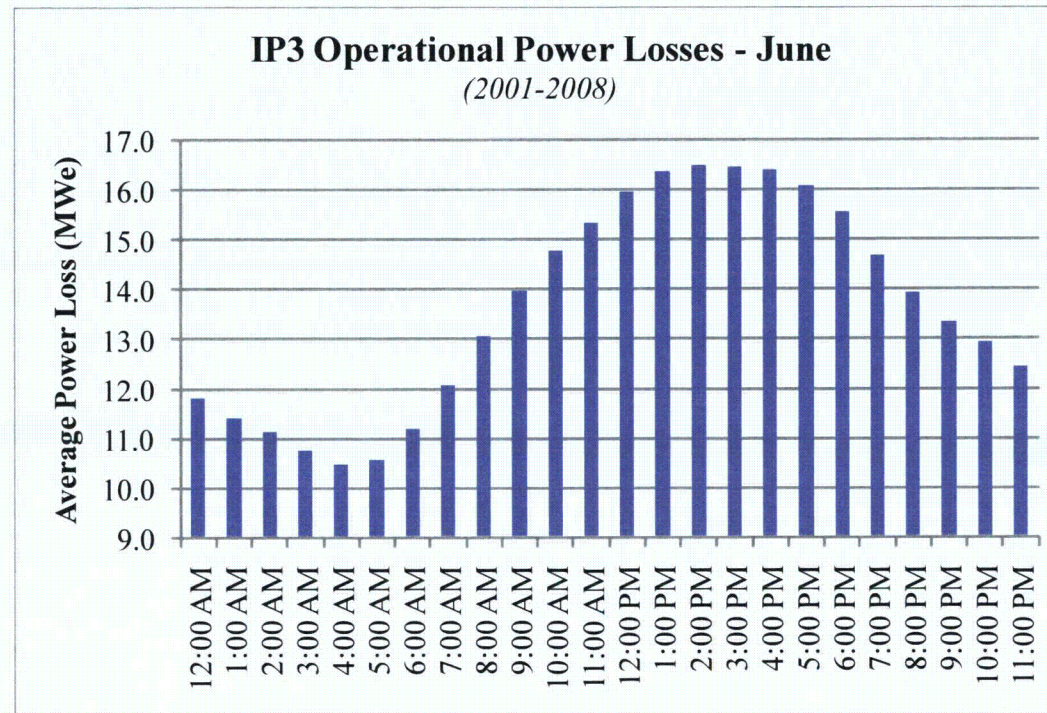
Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	2.4	19.0
1:00 AM	2.3	18.5
2:00 AM	2.2	18.1
3:00 AM	2.0	17.0
4:00 AM	1.9	16.9
5:00 AM	1.8	16.5
6:00 AM	1.9	17.4
7:00 AM	2.2	19.7
8:00 AM	2.6	21.4
9:00 AM	3.0	22.6
10:00 AM	3.5	23.0
11:00 AM	4.0	22.9
12:00 PM	4.3	23.5
1:00 PM	4.4	22.7
2:00 PM	4.6	24.1
3:00 PM	4.6	23.1
4:00 PM	4.5	22.1
5:00 PM	4.3	21.9
6:00 PM	3.9	22.1
7:00 PM	3.6	20.9
8:00 PM	3.3	20.2
9:00 PM	3.1	19.3
10:00 PM	2.8	18.7
11:00 PM	2.7	18.7



Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	7.9	22.8
1:00 AM	7.7	21.6
2:00 AM	7.5	20.9
3:00 AM	7.3	20.9
4:00 AM	7.1	21.0
5:00 AM	7.1	21.2
6:00 AM	7.4	22.2
7:00 AM	8.0	25.1
8:00 AM	8.8	27.6
9:00 AM	9.5	28.7
10:00 AM	10.2	29.4
11:00 AM	10.7	30.4
12:00 PM	11.2	29.8
1:00 PM	11.6	29.7
2:00 PM	11.7	29.6
3:00 PM	11.8	29.5
4:00 PM	11.7	29.1
5:00 PM	11.3	27.6
6:00 PM	10.7	25.7
7:00 PM	10.0	24.4
8:00 PM	9.5	23.9
9:00 PM	9.2	23.8
10:00 PM	8.8	23.7
11:00 PM	8.5	23.0

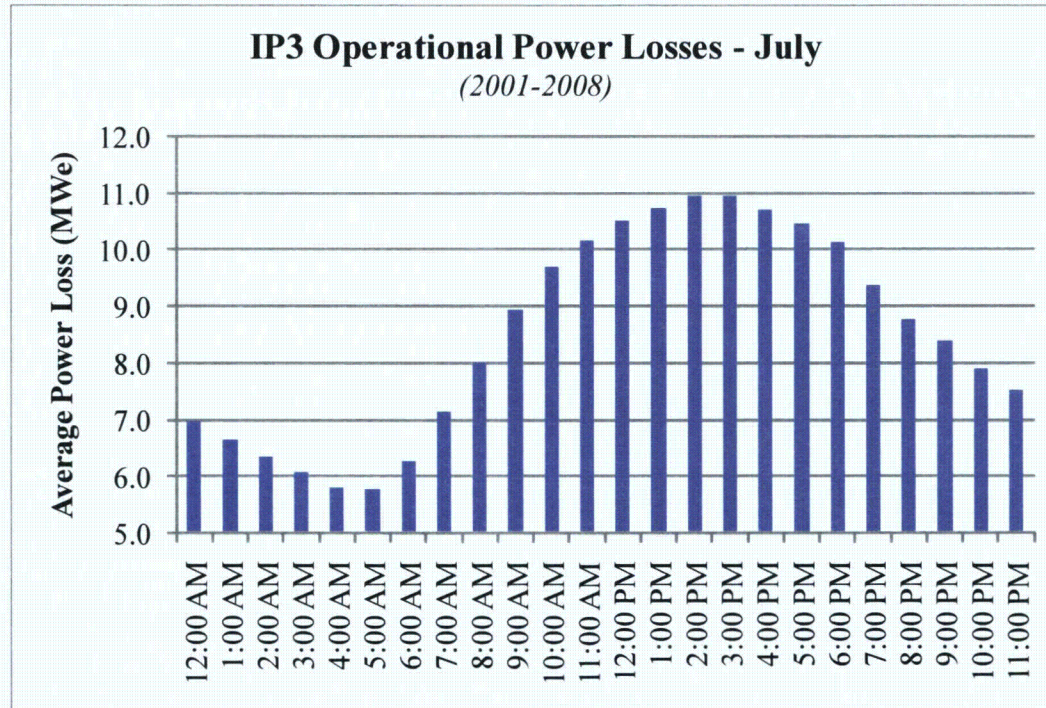


Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	11.8	27.9
1:00 AM	11.4	27.9
2:00 AM	11.1	27.5
3:00 AM	10.8	24.5
4:00 AM	10.5	24.3
5:00 AM	10.6	24.4
6:00 AM	11.2	26.5
7:00 AM	12.1	29.0
8:00 AM	13.1	30.7
9:00 AM	14.0	31.4
10:00 AM	14.8	33.2
11:00 AM	15.3	34.4
12:00 PM	15.9	35.5
1:00 PM	16.4	36.2
2:00 PM	16.5	38.0
3:00 PM	16.4	37.4
4:00 PM	16.4	36.0
5:00 PM	16.1	33.1
6:00 PM	15.6	33.3
7:00 PM	14.7	31.9
8:00 PM	13.9	30.6
9:00 PM	13.3	29.9
10:00 PM	12.9	28.8
11:00 PM	12.4	27.6

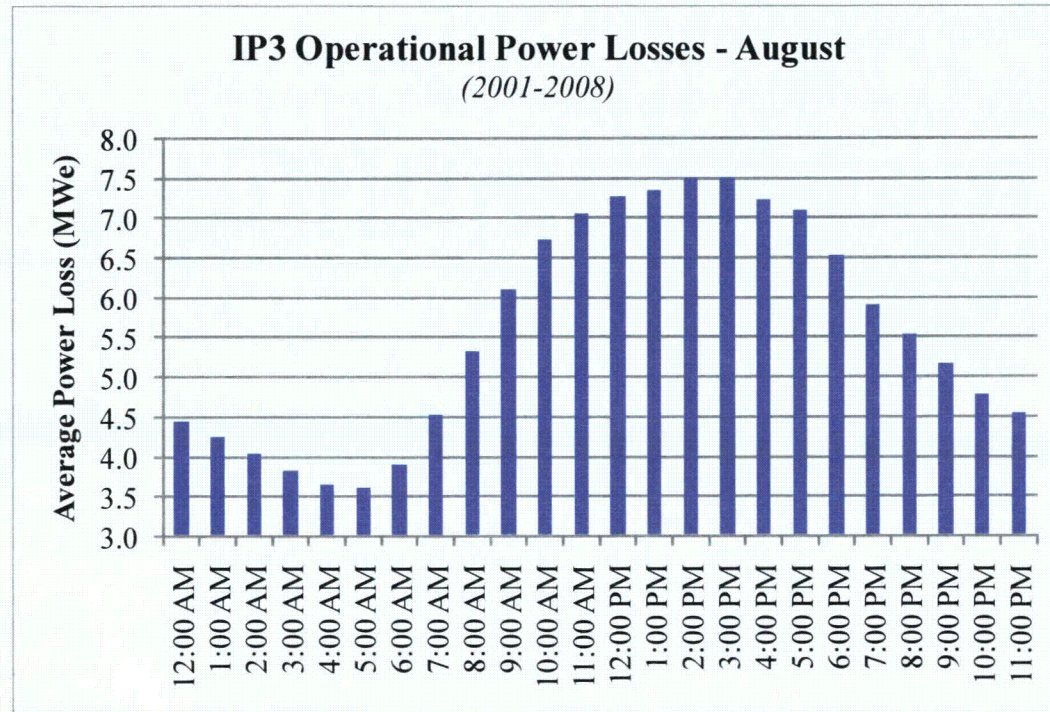




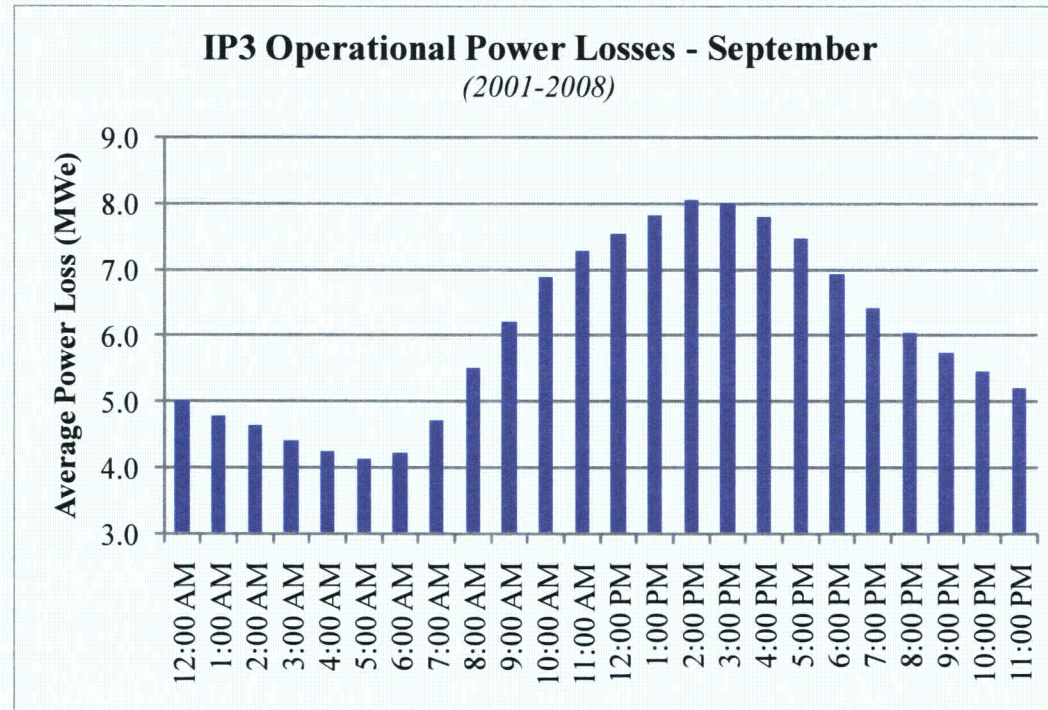
Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	7.0	26.9
1:00 AM	6.7	25.3
2:00 AM	6.3	24.6
3:00 AM	6.1	24.0
4:00 AM	5.8	23.1
5:00 AM	5.8	22.7
6:00 AM	6.3	23.8
7:00 AM	7.1	25.2
8:00 AM	8.0	26.5
9:00 AM	8.9	27.8
10:00 AM	9.7	28.6
11:00 AM	10.2	30.6
12:00 PM	10.5	31.3
1:00 PM	10.7	31.1
2:00 PM	11.0	31.0
3:00 PM	10.9	32.2
4:00 PM	10.7	32.0
5:00 PM	10.5	32.1
6:00 PM	10.1	32.2
7:00 PM	9.4	30.1
8:00 PM	8.8	28.4
9:00 PM	8.4	28.6
10:00 PM	7.9	29.1
11:00 PM	7.5	29.1



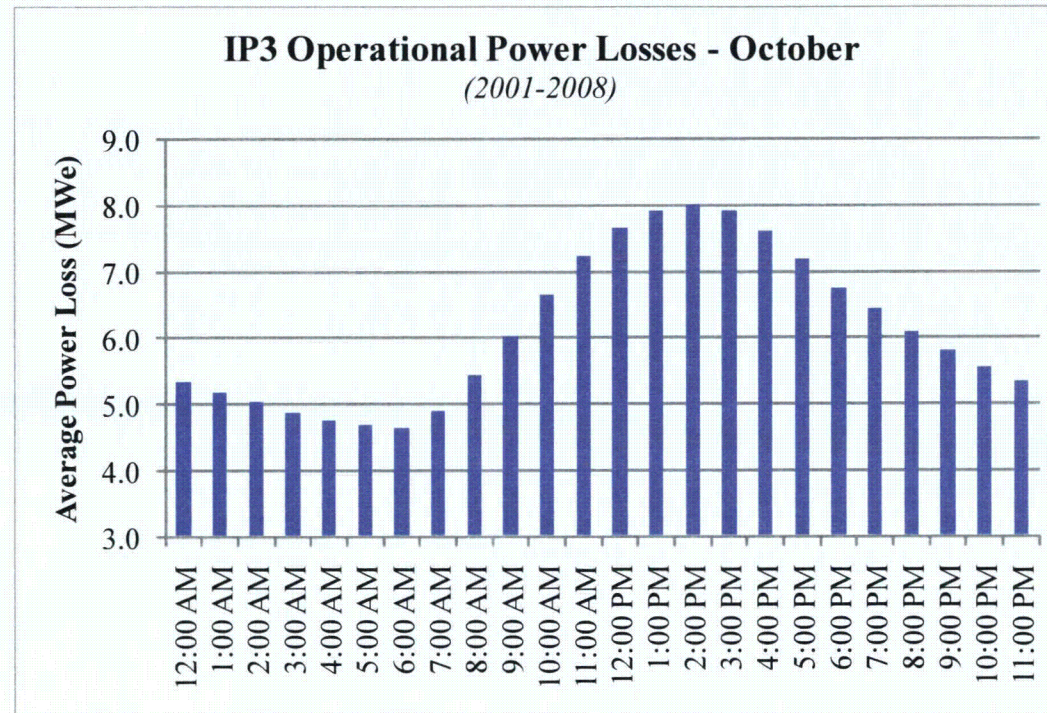
Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	4.5	21.0
1:00 AM	4.2	21.1
2:00 AM	4.0	20.3
3:00 AM	3.8	19.6
4:00 AM	3.7	19.0
5:00 AM	3.6	19.7
6:00 AM	3.9	20.6
7:00 AM	4.5	22.1
8:00 AM	5.3	23.9
9:00 AM	6.1	24.0
10:00 AM	6.7	25.6
11:00 AM	7.1	27.0
12:00 PM	7.3	27.5
1:00 PM	7.4	27.4
2:00 PM	7.5	27.6
3:00 PM	7.5	27.4
4:00 PM	7.2	27.1
5:00 PM	7.1	26.1
6:00 PM	6.5	23.5
7:00 PM	5.9	21.9
8:00 PM	5.5	21.8
9:00 PM	5.2	21.5
10:00 PM	4.8	21.8
11:00 PM	4.5	22.3



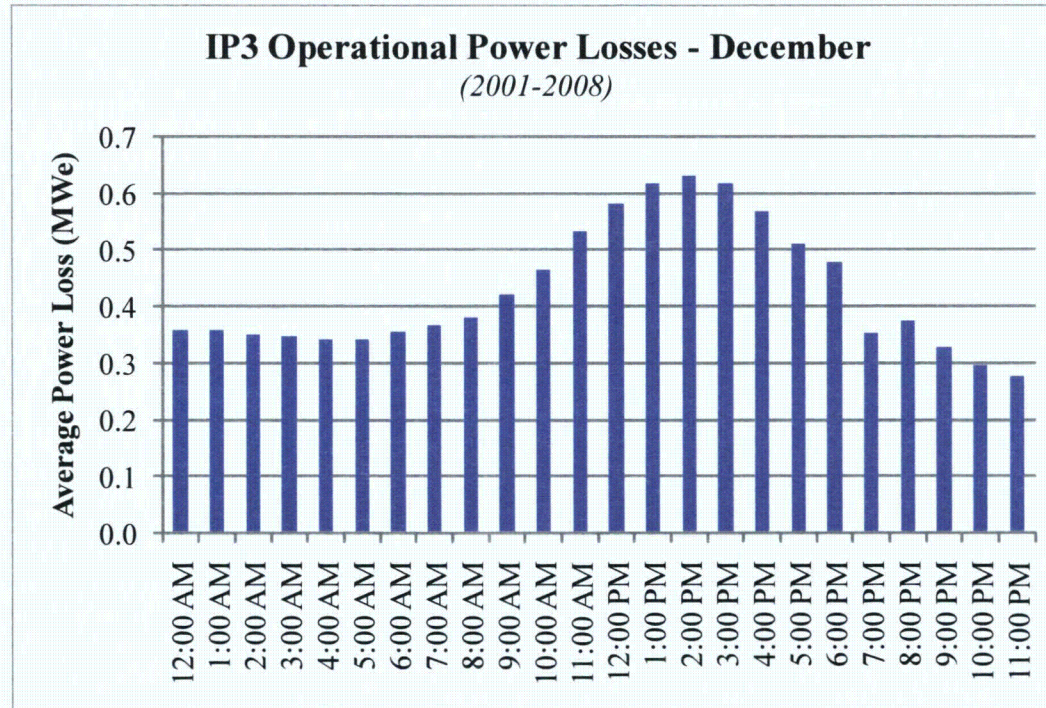
Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	5.0	21.1
1:00 AM	4.8	19.4
2:00 AM	4.6	21.3
3:00 AM	4.4	21.3
4:00 AM	4.2	20.2
5:00 AM	4.1	19.1
6:00 AM	4.2	19.8
7:00 AM	4.7	21.3
8:00 AM	5.5	21.5
9:00 AM	6.2	22.3
10:00 AM	6.9	22.2
11:00 AM	7.3	21.7
12:00 PM	7.5	23.3
1:00 PM	7.8	24.7
2:00 PM	8.1	25.1
3:00 PM	8.0	25.2
4:00 PM	7.8	23.9
5:00 PM	7.5	22.3
6:00 PM	6.9	20.9
7:00 PM	6.4	20.1
8:00 PM	6.1	20.3
9:00 PM	5.8	21.1
10:00 PM	5.5	21.3
11:00 PM	5.2	20.7



Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	5.3	20.6
1:00 AM	5.2	20.9
2:00 AM	5.0	20.6
3:00 AM	4.9	20.2
4:00 AM	4.8	20.3
5:00 AM	4.7	20.4
6:00 AM	4.6	20.7
7:00 AM	4.9	21.2
8:00 AM	5.4	21.8
9:00 AM	6.0	22.2
10:00 AM	6.7	23.0
11:00 AM	7.2	23.7
12:00 PM	7.7	23.3
1:00 PM	7.9	24.2
2:00 PM	8.0	24.9
3:00 PM	7.9	24.7
4:00 PM	7.6	24.1
5:00 PM	7.2	22.9
6:00 PM	6.8	23.0
7:00 PM	6.4	23.4
8:00 PM	6.1	23.2
9:00 PM	5.8	23.4
10:00 PM	5.6	19.9
11:00 PM	5.3	20.5



Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	0.4	14.2
1:00 AM	0.4	14.4
2:00 AM	0.3	14.4
3:00 AM	0.3	14.6
4:00 AM	0.3	14.4
5:00 AM	0.3	14.4
6:00 AM	0.4	15.8
7:00 AM	0.4	15.9
8:00 AM	0.4	16.0
9:00 AM	0.4	17.3
10:00 AM	0.5	16.2
11:00 AM	0.5	16.1
12:00 PM	0.6	16.2
1:00 PM	0.6	16.2
2:00 PM	0.6	16.0
3:00 PM	0.6	16.2
4:00 PM	0.6	16.9
5:00 PM	0.5	17.3
6:00 PM	0.5	20.1
7:00 PM	0.4	8.1
8:00 PM	0.4	12.2
9:00 PM	0.3	8.2
10:00 PM	0.3	7.0
11:00 PM	0.3	6.7



Time	Power Loss (MWe)	
	Average	Maximum
12:00 AM	0.4	14.2
1:00 AM	0.4	14.4
2:00 AM	0.3	14.4
3:00 AM	0.3	14.6
4:00 AM	0.3	14.4
5:00 AM	0.3	14.4
6:00 AM	0.4	15.8
7:00 AM	0.4	15.9
8:00 AM	0.4	16.0
9:00 AM	0.4	17.3
10:00 AM	0.5	16.2
11:00 AM	0.5	16.1
12:00 PM	0.6	16.2
1:00 PM	0.6	16.2
2:00 PM	0.6	16.0
3:00 PM	0.6	16.2
4:00 PM	0.6	16.9
5:00 PM	0.5	17.3
6:00 PM	0.5	20.1
7:00 PM	0.4	8.1
8:00 PM	0.4	12.2
9:00 PM	0.3	8.2
10:00 PM	0.3	7.0
11:00 PM	0.3	6.7

