# REFLECTION TRAVELTIME TOMOGRAPHY 

## AND THE

## MAXIMUM ENTROPY PRINCIPLE

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## LIST OF PUBLICATIONS

Whiting, P.M., 1991, Practical reflection tomography and maximum entropy imaging, 61st Ann. Internat. Mtg., Soc. Expl. Geophys., Expanded Abstracts, 986-989.

Whiting, P.M., 1991, Maximum entropy reflection tomography, Exploration Geophysics, 23, 459-464.

Whiting, P.M., 1993, Entropy and tomographic image reconstruction, Submitted to Geophysics.

Whiting, P.M., 1993, Practical reflection tomography for velocity field estimation, Submitted to Geophysics.

Whiting, P.M., 1993, Entropy and geophysical inverse problems, Submitted to Exploration Geophysics.

## SUMMARY

Conventional reflection tomography creates an estimate of subsurface seismic velocity structure by inverting a set of seismic traveltime data. This is achieved by solving a least-squares optimisation problem that finds the velocity and depth model that minimises the difference between raytraced and measured traveltimes. Obtaining the traveltime data can be difficult as manual picking of reflection times is required and all picked reflection events must be associated with the reflector depths defined in the model. Even with good traveltime data the optimisation problem is very non-linear and the surface restriction of the sources and receivers makes the problem generally underdetermined. These issues result in severe ambiguity and local minima problems.

This thesis shows that modifications to the conventional reflection tomography algorithm can make it a more practical and reliable procedure that is less likely to be trapped by local minima. The ray tracing procedure is changed so that reflector depths are not necessary and automatic traveltime interpretation can be successful. Entropy constraints are introduced (after being justified) which prevent unwarranted velocity structure from appearing. This feature adds significant stability and reduces the ambiguity problems. Staged smoothing of the optimisation function helps avoid local minima.

Synthetic data examples show that the algorithm can be very effective on noise free data. Adding noise to synthetic data reduces the algorithms effectiveness, but inversions of real data sets produces updated velocity fields that result in superior pre-stack depth migrations.

## §1 INTRODUCTION

Reflection seismic surveys are conducted with the aim of imaging subsurface geologic structures. Historically, seismic profiles have been used to aid detection of hydrocarbon reservoirs. These are most commonly found in sedimentary basins. Seismic profiles show reflection strength as a function of subsurface locations and acoustic traveltime. Ultimately, the explorationist requires an estimate of the actual depth structure.

Accurate depth imaging of seismic reflection data requires knowledge of the subsurface acoustic velocity field. During conventional seismic data processing, this subsurface velocity field is derived from CDP stacking velocities (see Telford et al., 1976) which are based on the normal (or hyperbolic) moveout assumption (see Taner and Koeler, 1969). The hyperbolic moveout assumption is only valid for velocity fields that do not vary appreciably in the lateral direction (hyperbolic moveout is exact only for constant velocity fields), but it has proved to be very successful as the sedimentary basins under investigation generally have little lateral variation on the scale of the recording aperture. Whenever significant lateral variations are present (for example, faulting, reef structures, channelling and low or high velocity zones), conventional processing can produce less than optimum results. In this case processing must be performed that is not limited by the hyperbolic moveout assumption.

The technique known as reflection tomography was demonstrated by Bishop et al. (1985). The aim of this technique is to derive the subsurface velocity field (in terms of interval velocities and depths) directly from the traveltimes of reflected seismic waves. No moveout assumption is made. The algorithm requires an array of reflected energy
traveltimes to be picked from the collected data, along with a set of reflector depths. The algorithm then iteratively updates the subsurface model in an attempt to minimise a constraint statistic which is basically the sum of the differences between the raytraced (modelled) traveltimes and the measured traveltimes. Extra terms are often added to the constraint statistic in an attempt to stabilise the inversion. The constraint statistic does not have to be based on traveltime residuals - the only requirement is that it is a useful measure of errors in the model.

Reflection tomography has not yet formed part of conventional processing. There are many reasons for this. Firstly, the required reflector depths are difficult to estimate and poor estimates can prohibit the inversion process from achieving useful results. Finding reflecting horizons can be a difficult task in itself - especially with complex structural geology (see Bording et al., 1987; Jacobs et al., 1992; Lailly et al., 1992). After estimating a set of reflectors, the traveltime data then have to be picked for energy reflected from these reflectors only. This is because the raytracing used in the inverse procedure (two-point raytracing), finds the traveltime between a surface source, a specified reflector and a surface receiver. Such restrictions imposed on the traveltime picking can make this step difficult and laborious (see Leger et al., 1989; Lailly et al., 1992).

Even when a reliable set of traveltimes and reflectors have been defined, a useful inversion result is not guaranteed. The surface restriction of the sources and receivers result in an inverse problem that is predominantly underdetermined. The iterative minimisation of the traveltime error is prone to entrapment by local minima and many areas of ambiguity exist (see van Trier, 1988). A solution to these problems must be found before reflection tomography will find more widespread use. An easily obtained,
reliable velocity field obtained from reflection tomography could be used to produce accurate seismic images in areas of complex structural geology.

The problems with reflection tomography, outlined above, have lead researchers to develop many related algorithms to extract velocity field information from the traveltimes of reflected energy (Biondi, 1988; van der Made and van Riel, 1988; Harlan, 1989; Sherwood, 1989; Toldi, 1989; Biondi, 1990). These algorithms have varying degrees of success and ease of use. Most of them simplify the problem in some way (for example, by defining velocity fields characterised by smooth spline functions only) in order to stabilise the inversion. Also, significant research has been conducted into the use of depth migration to simplify the traveltime data interpretation (van Trier, 1990; Harlan al. ${ }_{\star} \mathrm{t}_{\mathrm{K}}$, 1991a, 1991b; Jacobs et al., 1992). Bording et al., 1987, used depth migration to help ensure that the reflector depths were accurate, without providing any picking assistance.

The aim of this thesis is to overcome the problems of conventional reflection tomography without limiting its potential. An algorithm will be developed that is very simple to use and practical, yet has the potential for improving the image quality obtained from the conventional algorithm. The reflector restrictions imposed on traveltime interpretation will be removed. A staged smoothing procedure will stabilise the inversion, and entropy constraints will help solve ambiguity problems and add further stability.

Sword (1986) demonstrated the advantages of removing reflector depths from the model parameters of Bishop's reflection tomography algorithm. This requires a raytracing procedure that does not need defined reflector locations. When the apparent dip of a reflection is measured on common source and common receiver gathers, it is possible to estimate the intercept angles of the rays at the surface. With this information, raytracing can be performed by tracing from both the source and receiver locations into the model
until the two ends meet. Reflectors do not have to be defined but apparent dips must be measured along with the traveltime. The advantage of this method is that there are no restrictions on the reflections that can be picked. Automatic data picking routines are a practical approach in this case. Fault plane reflections, diffracted energy and multiple arrivals (see Lailly et al., 1992) all become available as valid data with this style of raytracing. These extra data categories can provide wider angular coverage and hence better solutions.

The use of smoothing in inverse problems is a well known method for stabilising an inversion (Biondi, 1988; Menke, 198 $\stackrel{4}{\prime}^{\prime}$, Williamson, 1990). It makes intuitive sense to use heavy smoothing early in the inversion, but to decrease the smoothing as the inversion improves the model - ultimately resulting in an unsmoothed velocity image. This type of approach initially determines the required low frequency velocity changes and gradually extracts the higher and higher frequency components of the modifications to the starting model. Williamson (1990) applied this type of smoothing approach to reflection tomography. Biondi (1988) used a similar approach in his beam-stack inversion. The algorithm which employs stages of decreasing smoothing, is effectively a means of damping the components of the inversion with small singular values (Jupp and Vozoff, 1975). Some local minima (caused by the small singular value components) are temporarily smoothed over, allowing the inversion to avoid them. The probability of finding the global minimum of the constraint statistic is increased.

Entropy, which plays a central role in this thesis, has a long history. Entropy constraints are commonly used to aid in the inference of a solution for undetermined inverse problems. Originally derived in the fields of thermodynamics (Jaynes, 1988a) and information theory (Shannon, 1948; Khinchin, 1957), entropy has recently been applied
in many other disciplines of science; for example, image processing, astronomy and geophysics (Reitch, 1977; Skilling and Gull, 1985; Bassrei, 1990; Whiting, 1991b). Maximum entropy solutions to inverse problems have the special property that they do not contain any structure other than that needed to obtain a solution. The maximum entropy solution is unique and, to the extent that the data is reliable, the solution is maximally unbiased in terms of structure. Entropy is maximised at every iteration, ensuring that no unjustifiable decisions are made at any stage of the iterative inversion. Since spurious anomalies are prohibited, maximising entropy has a stabilising effect on the inversion. However, it can be difficult to associate the current reflection tomography problem with applications of the maximum entropy principle within other disciplines. But with the aid of the notion of contingency tables and simple tomographic problems, an instructive association can be made (see Chapter 3).

With entropy constraints, the inverse problem becomes the simultaneous minimisation of a constraint statistic (error function) and the maximisation of the entropy function. The Gauss-Newton method of inversion used by Bishop et al. (1985) is replaced by a method using local quadratic approximations to the constraint and entropy functions within a 4 or 6 dimensional subspace of the original model space (with typically thousands of dimensions; see Kennett and Williamson, 1988, for more on subspace methods). Such an inversion algorithm was described by Skilling and Bryan (1984). Suitable modifications are necessary to incorporate stages of decreasing smoothing. The inversion process can also continuously estimate the extent of validity of the local quadratic approximations, and adjust the iteration dependent distance limit accordingly. This ensures the most rapid inversion possible without producing instability. The constraint statistic is the sum of the squared errors which are weighted to approximate an L1 norm so that erroneous data do
not have a detrimental effect on the inversion.
Synthetic data examples show that significant velocity anomalies can be resolved with the modified reflection tomography algorithm. The maximum entropy images obtained clearly have less spurious structure - the background noise ${ }^{1}$ is suppressed. The strategy of decreasing smoothing successfully avoids some local minima problems.

When real seismic data is used, the interpreted traveltime data will be inherently noisy. By adding noise to some synthetic data it was found that $10 \%$ noise (added to the measured angles) was tolerable whereas $20 \%$ noise may restrict any useful velocity updates from being achieved. Reflection tomographic inversions of two real data sets show that significant improvements can be achieved in real situations. The validity of these updates is demonstrated by improved pre-stack depth migration results.

The spatial frequency of velocity variations obtainable from conventional means is low compared to that possible with reflection tomography. The most obvious role for reflection tomography is to verify and extend the best velocity model obtained by conventional means. With the modified reflection tomography algorithm proposed in this thesis, such updated velocity models are more reliable and obtained with little manual intervention.

Other important works relevant to this thesis are (alphabetical order) Ables (1974), Al-Yahya (1989), Berkhout (1987), Bracewell (1978), Burg (1967), Carrion et al. (1993), Catlin (1989), Chiu and Stewart (1987), Cornwell (1983), Delprat and Lailly (1990), de Vries and Berkhout (1984), Dusaussoy and Abdou (1991), Etgen (1988), Farra and Madariaga (1988), Finn and Backus (1991), Frieden (1972), Gill and Murray (1974),

[^0]Good (1963), Gull (1988), Gull and Daniell (1978), Gull and Skilling (1983), Jackson (1979), Jaynes (1957, 1982, 1984, 1985, 1988b), Justice et al. (1989), Kennett (1988), Kennett et al. (1988), Kennett and Harding (1985), LaBrecque (1990), Lailly et al. (1990), Landa et al. (1989), Langan et al. (1985), Lines (1991), Lynn and Claerbout (1982), Marquardt (1963), McGillivray and Oldenburg (1990), Michelena and Harris (1991), Moser (1991), Oldenburg and Ellis (1991), Phillips and Fehler (1991), Press et al. (1989), Reitsch (1977, 1985), Scheuer and Oldenburg (1988), Shaw and Orcutt (1985), Shore (1984), Singh and Singh (1991), Skilling (1988), Skilling and Gull (1984), Smith et al. (1984), Stewart (1989), Stork (1992), Stork and Clayton (1991), Sword (1987), Tarantola (1987), Tikhonov and Arsenin (1977), Ulrych et al. (1990), Vidale (1989), White (1991) and Zhou et al. (1992).

## §2 REFLECTION TRAVELTIME TOMOGRAPHY - PRACTICALITIES

### 2.1 Ray tracing with triangular cells

The development of a traveltime tomography algorithm begins with consideration of ray tracing routines. Ray tracing can easily be the most time consuming component of the overall algorithm and generally a trade-off between accuracy and speed is considered. One purpose of the current study is to improve reflection tomography in general, and rather than create a possible source of error with inaccurate ray tracing, computational speed was sacrificed for the sake of accuracy.

Bishop et al. (1985) used a ray tracing procedure as described by Langan et al. (1985). This procedure divides the image into square cells and computes approximate linear velocity functions of the form $V=a+b x+c z$ within each cell ( $a, b, c$ constants). The four nodes (see Figure 2.1-1) at the corners of the square cell will have velocities that do not generally satisfy such a linear relationship. However, a simple velocity structure is desired for rapid computation. So a linear velocity field is


Fig. 2.1-1 Four velocity nodes are used to construct a square cell. A typical circular raypath is shown.
approximated to fit the four corner velocities, causing velocity discontinuities at the boundaries of the cells. Within the locally linear velocity fields, the raypaths are either circular or straight and the velocity discontinuities cause refraction at the boundaries. Langan et al. (1985) made further approximations to simplify the calculations of the circular arcs. The combination of the approximations used here may be insignificant, but for the present study any such possible source of error is to be avoided at the outset. More accurate ray tracing can be achieved with triangular cells.

The use of triangular instead of square cells has significant initial difficulties due to the existence of a sloping side (see Figure 2.1-2). When searching for raypath intersections with the sides of the cells, considerably more book keeping has to be done to cope with the sloping sides. However, there are significant advantages as well. Each triangular cell is defined by only three velocity nodes, uniquely specifying a linear velocity field. Therefore all velocities are continuous at cell boundaries, and each velocity node is honoured exactly. The resulting rays are circular within each cell, without refraction at the boundaries (see Figure 2.1-2). It is also intuitively easier to fit a triangular mesh to a surface than to try and fit a square mesh.

The ray tracing procedure was


Fig. 2.1-2 The geometry of the triangular cell configuration and some typical circular arcs.
designed with a kernel program capable
of tracing through any given triangular cell. This kernel program, named RAYTRACE
can be found in the appendix. The following is a summary of the flow of this program.
i) The input data are the spatial locations and velocities of the three grid nodes defining the triangular cell, along with the entry location and entry angle of the ray.
ii) Compute lateral and vertical velocity gradients by finite differencing across the corners of the triangle.
iii) Compute the velocity at the input ray location.
iv) Determine the angle between the velocity gradient vector and the reference axes and the magnitude of the gradient.
v) If the gradient is zero, flag the ray for straight ray tracing within this cell.
vi) Compute the radius of curvature for the ray's circular arc within this cell.
vii) If the radius is larger than a threshold value, flag the ray for straight ray tracing ${ }^{1}$. For the purposes of raytracing efficiently the threshold needs to be as low as possible, but surprisingly significant errors are easily created.
viii) Compute the centre of the circular arc.

Particular care must be taken to ensure that the circular arc is curving in the correct direction. That is, the centre must be chosen on the correct side of the ray ${ }^{2}$.

[^1]ix) Intersect the circular arc with the sides of the triangular cell (see Figure 2.1-3), thus determining all possible exit points.
x) Determine the exit angle for each of the possible exit points ${ }^{3}$.
xi) Compute the arc length from the entry point to each possible exit point around the circle (in the direction of energy transfer).
xii) The exit point with the shortest possible arc length is the required exit point.

The accuracy and integrity of this ray tracing code is of paramount importance for the traveltime inversion algorithm. Thus great care was taken to ensure that all possible ray configurations were correctly handled.

Most of the calculations performed by the subroutine RAYTRACE are performed with double precision calculations to avoid rounding errors. Originally, a threshold of $10^{4}$ metres was set for circular rays. However, a serious problem was uncovered while analysing constraint statistic derivatives (see section 2.4). It was found that this was not accurate enough and that the threshold had to be increased to a radius of $10^{6}$ metres.


Fig. 2.1-3 The arrow represents the entry location and orientation of the ray. The dashed curve is the circular arc that the ray must traverse. A, B and C are possible exit points.

[^2]One potential problem with the ray tracing scheme as defined above, is an artificially introduced asymmetry. As can be seen from the example in Figure 2.1-2, the four velocity nodes can be split into triangles with a diagonal line sloping upwards to the right, or equivalently, upwards to the left. Irrespective of the sense of triangulation, such cells can make symmetric velocity models slightly asymmetric.

However, as long as the cell size is small compared to the features being resolved, the induced asymmetry will be negligible. ${ }^{4}$

The ray tracing algorithm described above is computationally intense. Langan et al., 1985, go to considerable lengths to avoid the evaluation of trigonometric functions. No such compromises were made in the studies discussed in this thesis.

There are other approaches to the ray tracing problem which may prove adequate for reflection tomography applications, while possibly being more efficient. These approaches include finite difference traveltime modelling (eg. Vidale, 1989) and shortest path calculations (Moser, 1991). Currently, the ray tracing portion of the reflection tomography algorithm is the most time consuming component, taking up to $80 \%$ of the total computation time (depending on the size of the model; see section 4).

[^3]2.2 Removal of reflectors from problem parameterisation.

The conventional reflection tomography algorithm, as introduced by Bishop et al., 1985, requires a set of specific reflectors to be chosen and in turn, that all traveltime data picked must be for energy that has reflected from one of these reflectors. These restrictions were imposed to enable two-point ray tracing to compute theoretical travel times between known source and receiver locations. However, these restrictions introduce a lot of practical difficulties into the conventional algorithm. Firstly, it is often difficult to define an adequate set of reflectors in areas where reflection tomography could be useful (eg. structurally complex areas). It is very difficult to ensure that travel times for these reflectors are correctly interpreted from the pre-stack gathers. A further difficulty is that these chosen reflectors must have depths assigned before the tomographic inversion can begin. If no well data is available it is not easy to assign accurate depths. Leger et al., 1989, gave a graphic description of how difficult the travel time interpretation can be; they had to repick the data many times in order to get the correct inversion (luckily, they knew the type of results required). Finn and Backus (1991) made similar comments regarding the difficulty of traveltime picking during their presentation. They found the picking so difficult, they stopped picking travel time data for one corner of their prospect. Lailly et al. (1992) described how difficult the traveltime interpretation can be in areas with complex structural geology.

A lot of research effort in recent years has been aimed at reducing the effort required to extract the traveltime information from the data and prepare the initial
model. Here it is assumed that any algorithm designed to estimate the subsurface interval velocity field from the travel times of reflected seismic energy, is a form of reflection tomography. Along these lines, many researchers have developed algorithms to achieve the results of conventional reflection tomography, but with the emphasis on the ease of application.

Biondi $(1988,1990)$ developed an algorithm for subsurface interval velocity estimation with the help of "beam-stacks". The purpose of using beam-stacks was to remove the need to pick the traveltime data. An inversion procedure was set up to find the interval velocity distribution that maximised the resulting beam-stack energy. Beam-stacking is essentially a localised slant-stacking process across a gather of traces. Because it is localised, it is capable of resolving the non-hyperbolic moveout associated with complex velocity fields. Since a small group of traces are beam-stacked, the method is less troubled by noisy data, but its resolution is slightly decreased. After performing the beam-stacks, it is possible to find the velocity model that maximises this beam-stack energy. This velocity model is effectively a tomographically derived velocity field, and manual data picking was not necessary.

A somewhat similar approach was made by Toldi, 1989. He used CMP stacking semblance to derive an interval velocity model without the need for picking any data. This algorithm simplifies the practical aspects of reflection tomography, but its range of application is reduced. Given a proposed interval velocity model, Toldi used computed stacking velocities to evaluate the overall stacking semblance of the data (similar to overall beam-stack energy). Iterative changes to the velocity model were made to maximise the overall semblance. The technique is restricted by the use of stacking semblance values which are based on the hyperbolic moveout assumption.

Success can only be expected for velocity models that are smooth enough to produce roughly hyperbolic moveout.

Picking the required traveltime data after pre-stack depth migration has been proposed to simplify the conventional reflection tomography algorithm (Van Trier, 1990; Harlan et al., 1991a; Jacobs et al., 1992; Lailly et al., 1992; Stork, 1992). These researchers have acknowledged that traveltime data picking is very difficult and laborious, but they have noticed that it becomes easier if the data has been pre-stack depth migrated with an approximate velocity field. When picking the data from common-offset gathers, the reflection events are easier to interpret after an approximate migration. Accommodating traveltime data after pre-stack depth migration requires modifications to the procedure of Bishop et al., 1985, primarily for Frechet derivative computation ${ }^{1}$. In general, this type of algorithm has the computationally expensive pre-stack depth migration process between each iteration. The bulk of this expense can be avoided with approximate methods of residual migration (Van Trier, 1990). Traveltime data picking has not been avoided but it has been made easier.

Many other simplifications to the conventional reflection tomography algorithm (Bishop et al., 1985) have been published. Van der Made and Van Riel (1988) used stacking velocity values and zero-offset traveltimes to invert for the subsurface interval velocity field. Using stacking velocities without offset-dependent traveltimes is an extreme case of avoiding pre-stack data picking. Similarly, Sherwood (1989) generated hyperbolic traveltime data directly from stacking velocities. These methods are simple to use but they have difficulties in areas of complex velocity structure. Harlan

[^4]$(1989,1991 b)$ simplified the data picking by incorporating stacking velocities and simplifying the data set to be interpreted. These research efforts demonstrate the difficulty of traveltime data picking for the conventional reflection tomography algorithm.

Two-point raytracing is used to compute theoretical traveltimes in the conventional reflection tomography algorithm. This approach to raytracing requires specifically defined reflectors and a fan of rays to be traced for every shot and each reflector (Figure 2.2-1). This is why the reflector depths must be specified as parameters of the inversion model - however, they are not known and must be allowed to vary. These extra parameters complicate the problem of determining the interval velocity field, introducing issues like velocity/reflector depth trade-off. Complications such as these are avoided when it is realised that two-point raytracing is not essential.

Conventional reflection seismic data is recorded in gathers of closely spaced traces. The array of receivers makes it possible to determine the angle of incidence of


Fig. 2.2-1 Two-point raytracing finds the raypath between a source, receiver and a given reflector. A fan of rays is shot with interpolation to the actual receiver location.
the seismic energy with the surface. If the near surface velocity is known, then the apparent dip of some reflected energy on a common-source gather is all that is required to compute the emergence angle of the energy at the receivers (see Figure 2.2-2). Similarly, the apparent dip of the same energy on a common-receiver gather gives the impingement angle at the source array. Having both surface angles, it is possible to trace both ends of the ray back into the model until they meet (Figure 2.22). Reflected traveltimes are computed without any defined reflectors. This approach to raytracing will be called two-end raytracing in contrast to two-point raytracing.

Sword (1986) made use of two-end raytracing in reflection tomography. He also observed that errors associated with estimating the surface angles can be detrimental to the inversion procedure. Since the angles are given by dips measured from the data, there will always be some measurement error. Sword (1986) found that such errors could make the estimated traveltimes unstable - especially if the source and receiver are relatively close together (see Figure 2.3-3). Fortunately, Sword discovered another statistic, $X_{e r r}$, which is more stable than the traveltime. This new statistic is the spatial separation of the two ends of the ray at the depth $Z$ where the total traveltime equals the actual traveltime. If the velocity model is correct, the two ends of the ray must meet at this depth and $X_{e r r}$ will be zero. This statistic makes it possible to use two-end ray tracing in reflection tomography.

The $X_{e r r}$ statistic was compared to the traveltime statistic in a simple synthetic reflection tomography experiment. The actual velocity model comprised grid points with a velocity of $2000 \mathrm{~m} / \mathrm{s}$, except for one anomalous point of $2050 \mathrm{~m} / \mathrm{s}$. A set of synthetic rays were generated in a symmetric pattern over the anomalous velocity.


Fig. 2.2-2 The principle behind two-end raytracing.


Fig. 2.2-3 The $X_{e r r}$ statistic is more stable than traveltime.

Figure 2.2-4 shows the results of an inversion which used traveltimes as the constraint statistic. This inversion has performed poorly. Conversely, Figure 2.2-5 displays the successful results of the inversion of the same problem with the use of $X_{e r r}$ as the constraint statistic.


Fig. 2.2-4 Inversion results using traveltimes. 43 rays were used. The marked velocity should be $2050 \mathrm{~m} / \mathrm{s}$ ( $2000 \mathrm{~m} / \mathrm{s}$ for the rest). (See the footnote on page 15)

|  |  |  |  |  |  |  |  |  |  |  |  |
| :--- | :--- | :--- | :--- | :--- | :--- | :--- | :--- | :--- | :--- | :--- | :--- |
|  |  |  |  |  |  |  |  |  |  |  |  |
| 2003 | 1999 | 2004 | 2004 | 1998 | 2002 | 1998 | 2004 | 2000 | 2001 | 2003 |  |
| 2000 | 2000 | 2012 | 2000 | 1997 | 2000 | 1996 | 2001 | 2007 | 2003 | 2001 |  |
| 2000 | 1999 | 2002 | 2004 | 1998 | 1991 | 1996 | 2009 | 2010 | 1999 | 2000 |  |
| 2000 | 2000 | 2006 | 1995 | 2010 | 1990 | 2004 | 2008 | 2003 | 2000 | 2000. |  |
| 2000 | 2000 | 2008 | 2010 | 2001 | 2042 | 1997 | 2006 | 2006 | 2000 | 2000 |  |
| 2000 | 2000 | 2005 | 2017 | 2011 | 1994 | 1992 | 2005 | 2003 | 2000 | 2000. |  |
| 2000 | 2000 | 2000 | 2005 | 2001 | 1994 | 1992 | 1988 | 1998 | 2000 | 2000 |  |
| 2000 | 2000 | 1997 | 1996 | 1998 | 1997 | 1999 | 1994 | 1996 | 2000 | 2000 |  |
| 2000 | 2000 | 1999 | 1996 | 1999 | 1998 | 2000 | 1999 | 1998 | 2000 | 2000 |  |
| 2000 | 2000 | 2000 | 1999 | 2000 | 2000 | 2000 | 2000 | 2000 | 2000 | 2000 |  |
| 2000 | 2000 | 2000 | 2000 | 2000 | 2000 | 2000 | 2000 | 2000 | 2000 | 2000 |  |
|  |  |  |  |  |  |  |  |  |  |  |  |

Fig. 2.2-5 An identical inversion to that of Fig. 2.2-4 except that $X_{\text {err }}$ 's were used instead of traveltimes. (See the footnote on page 15)

It has been claimed that this method of raytracing, without reflectors, will suffer because it does not force all reflections from a single reflector to focus at the same depth (Sword, 1986). The depth location of the ends of two adjacent rays that reflect from the same reflector, may be slightly different due to errors in the surface angle estimation, for example. Conventional reflection tomography forces rays like these to reflect at the same depth. However, it is not clear whether this reflector constraint does not cause more problems than it solves. More difficult picking and velocity/reflector depth trade-off problems certainly appear to be very substantial problems.

An advantage of using two-end raytracing is the increased number of data that can be picked. Without pre-defined reflectors it is a simple matter to include any reflected/scattered energy as data, even from very small reflectors. Diffracted energy as well as energy that has reflected from fault planes are now naturally admissible. The result is more data with a greater range of angular coverage (from diffracted/fault plane energy), both of which will enhance the results of the inversion process. The extra rays that can be defined may well compensate for the focussing issues described earlier.

Two-end raytracing can also avoid the expenses of two-point raytracing. For accurate two-point raytracing, more rays have to be traced than there are receivers. That is, for each required raypath, more than one ray is traced because the final surface location of the ray cannot be predicted. The required traveltime is interpolated from the traced times. With two-end raytracing, exactly one ray is traced for each required raypath. A further saving is afforded since no reflection logic is required.

### 2.3 Automatic data picking using complex attributes

Conventional reflection tomography requires specific events to be interpreted and careful, manual picking is a major and laborious step. Interactive workstation software has been proposed to make this step less arduous (Harlan et al., 1991a; Harlan et al., 1991b). However, two-end raytracing makes it possible to let a computer do all the work. Two-end raytracing (see section 2.2 ) makes traveltime data picking virtually unrestricted. Any reflected energy ${ }^{1}$ can be picked as long as it is coherent enough to enable its dip to be estimated (on shot and receiver gathers). Automatic data picking is potentially simpler and more successful.

Sword (1986) used localised slant-stacks to identify reflection events and estimate their apparent dip. This approach applies a series of time shifts to a small number of traces from a common source or receiver gather, followed by summation. This slant-stacking process transforms the small group of traces from the time/offset ( $t-x$ ) domain to the centre trace intercept/ray parameter ${ }^{2}(\tau-p)$ domain;

$$
\hat{y}\left(\tau, p_{j}\right)-\sum_{i-\frac{-(N-1)}{2}}^{\frac{(N-1)}{2}} y\left(\tau-i p_{j} \Delta x, i \Delta x\right)
$$

where
$y(t, x)$ is the original data set,
$t$ is the travel time,
$x$ is the offset from the centre trace of the group,

[^5]$N$ is the number of traces in the group (odd),
$\tau$ is the centre trace intercept time,
$p_{j}$ is the ray parameter equivalent to one of the $j$ dips being searched for, and,
$\Delta x$ is the spatial separation of the traces.
Once the data have been transformed in this way, the $\tau-p$ traces are searched for amplitude maxima. Each maximum defines a time on the centre trace and an apparent dip. These results are assigned to the centre trace location and the group of traces is rolled along by one trace, and a new $\tau-p$ transform is performed.

Once all traces of every common shot gather have been transformed and picked, the data are re-ordered into common receiver gathers. The $\tau-p$ transformation and picking process is repeated on each common receiver gather. In this way, both the apparent shot and receiver domain dips are obtained, as required for two-end raytracing. This process is successful (Sword, 1987), but it is also very computationally intense.

In 1988, Scheuer and Oldenburg described an algorithm that is capable of extracting the same information as achieved with localised slant stacks. This technique uses the complex attributes of the seismic traces. It is a simple process to compute the instantaneous frequency and instantaneous wavenumber at any point on a gather, and then take their ratio to give the apparent dip at that point. Obtaining the instantaneous attributes requires a single fast Fourier transform and its inverse for each trace. Such an algorithm will require considerably less computational effort than the previously described slant stack approach.

Complex trace attributes are derived from relatively simple theory. It can be assumed that any recorded trace is the real projection of some complex trace. Any imaginary component can be assigned to create such a trace, but only one particular imaginary component retains the characteristics of the original trace. Since the recorded trace is real and causal (zero amplitude for $t<0$ ), its Fourier transform is hermitian. Either the positive or negative half of such a Fourier spectrum constitutes enough information to recover the original total spectrum. It will be shown that zeroing the negative frequencies creates a complex trace with an imaginary part that is the Hilbert transform of the real part (the original trace). Such a complex trace allows for the evaluation of instantaneous attributes (see Bracewell, 1978).

If the recorded seismic trace is $x(t)$ then the complex trace, commonly referred to as the analytic signal, is

$$
a(t)=x(t)+i \hat{x}(t)
$$

where $\hat{x}(t)$ is the Hilbert transform of $x(t)$. The time domain Hilbert transform is defined as the time domain convolution with a digital filter $h(t)$ whose Fourier transform is $H(\omega)=-i \operatorname{sgn} \omega$. It is usually more efficient to apply the Hilbert transform in the frequency domain. Consider the Fourier transform of $x(t)$,

$$
X(\omega)-\int_{-\infty}^{\infty} x(t) e^{i \omega t} \mathrm{~d} t
$$

The negative frequencies can be suppressed by multiplying $X(\omega)$ by the unit step function ${ }^{3}, U(\omega)$, and then

[^6]$$
f(t)=\frac{1}{2 \pi} \int_{-\infty}^{\infty} U(\omega) \cdot X(\omega) e^{-i \omega t} \mathrm{~d} \omega
$$
is the inverse Fourier transform. The inverse Fourier transform of $U(\omega)$ is $u(t)-\frac{1}{2} \delta(t)+\frac{1}{i 2 \pi t}$ (see Berkhout, 1987), where $\delta(t)$ is the Dirac delta function. The convolution theorem gives ${ }^{4}$,
\[

$$
\begin{aligned}
f(t) & =\left(\frac{1}{2} \delta(t)+\frac{1}{i 2 \pi t}\right) * x(t) \\
& =\frac{1}{2} x(t)+\frac{1}{2} i h(t) * x(t) \\
& =\frac{1}{2}(x(t)+i \hat{x}(t)) .
\end{aligned}
$$
\]

Therefore, $f(t)=\frac{1}{2} a(t)$. In summary, the analytic signal is easily obtained by Fourier
transformation, zeroing the negative frequency components and inverse Fourier transformation.

The analytic signal, or complex trace, allows the trace to be viewed as a phasor rotating about an axis (see Bracewell, 1978). The length of the phasor is the instantaneous amplitude and its rate of rotation gives the instantaneous frequency. The real projection of this complex trace is the recorded signal, $x(t)$, and the imaginary projection is $\hat{x}(t)$, known as the quadrature trace. The phasor is never allowed to rotate
${ }^{4}$ Note that the Fourier transform of $H(\omega)--i \operatorname{sgn} \omega$ is $h(t)-\frac{-1}{\pi t}$.
backwards ${ }^{5}$ - frequency is always positive. These attributes of the complex seismic trace allow for the rates of change to be evaluated in both the temporal and spatial dimensions, and can therefore be used for the computation of the local dip of the dominant plane wave energy.

The subroutine PICKSPC5 (see the appendix) was coded to compute the complex attributes for a common source gather and detect the apparent dips of the data within the gather. A separate subroutine, PICKCRP5 (also see the appendix), uses the picks made in the common source domain and extracts the corresponding apparent dips in the common receiver domain. The algorithm used by PICKSPC5 is outlined below.
a) Read a common source gather of traces.
b) Convert each trace to the frequency domain (one dimensional FFT's).
c) Zero all the negative frequency components.
d) Transform all traces back to the time domain.
e) Compute instantaneous amplitudes $\left(\sqrt{x^{2}(t)+\hat{x}^{2}(t)}=A(t)\right)$.
f) Pick maxima of instantaneous amplitudes (not too close to each other).
g) Compute instantaneous phases $\left(\tan ^{-1}\left\{\frac{\hat{x}(t)}{x(t)}\right\}-\phi(t)\right)$.
h) Extract eight nearest neighbour instantaneous phases around each maximum of instantaneous amplitude (see Figure 2.3-1).
i) Compute instantaneous frequencies for each of the three temporal triplets. Do this in two halves. For example, around point 2 (see Figure 2.3-1) compute $\left(\phi_{3}-\phi_{2}\right)$

[^7]

Fig. 2.3-1 The nine instantaneous phase points used to compute the apparent dip of the amplitude maxima at node 5 .
and $\left(\phi_{2}-\phi_{1}\right)$ and compare. If these values are quite different, declare the pick made at point 5 unstable, and discard it. Otherwise, average the two differences and divide by $\Delta t$ to give the instantaneous frequency.
j) If the three estimated instantaneous frequencies are similar, average them.
k) Similarly compute the apparent wavenumber by using three consecutive horizontal points (see Figure 2.3-1). Note that the instantaneous wavenumber can be negative. It is this property that provides the sense of the apparent dip.

1) Compute the apparent velocity by dividing the computed instantaneous frequency by the apparent wavenumber.
$\mathrm{m})$ Convert this to dip using the known surface velocity, $V_{\text {SURF }}$,

$$
D I P=\sin ^{-1}\left(\frac{V_{\text {SURF }}}{V_{A P P A R E N T}}\right)
$$

n) Output this dip angle with the time of the instantaneous amplitude maximum
and the source/receiver locations to a file. This file will be used by PICKCRP5 (see the appendix).
o) Further consistency checks may be included before output. Such checks may, for example, only output if similar picks have been made nearby. Such extra checks are not always helpful.

Figure 2.3-2 shows the results of this algorithm when applied on some synthetic data. The picks seem to be generally representative of the data on the input gather.

All the initial testing of this complex attribute picking algorithm was performed on the synthetic data shown in Figure 2.3-2. Table 2.3-1 shows that the measured dips


Fig 2.3-2 The dip angle picks made by the complex attribute algorithm. The picks are displayed as small dip bars centred on the pick location.
were clustered into groups of similar values separated by sharp jumps, whereas regularly changing dips were expected. This problem was eventually traced to the nearest millisecond accuracy of the synthetic trace generation program. It was surprising that nearest millisecond rounding was clearly observed by the complex attribute picker. The accuracy of the algorithm promises to be high. Table 2.3-1 also records another potential problem since the picked maximum amplitude points are output to the nearest sample period (4 milliseconds in this case). More on this later.

When the complex attribute picker is applied to real data, many more mispicks

| Time | Measured Dip <br> (radians) | Theoretical Dip <br> (radians) |
| :--- | :--- | :--- |
| 0.324 | 0.5956 | 0.5667 |
| 0.320 | 0.5005 | 0.5504 |
| 0.316 | 0.5895 | 0.5337 |
| 0.312 | 0.5915 | 0.5167 |
| 0.312 | 0.4126 | 0.4993 |
| 0.308 | 0.4125 | 0.4817 |
| 0.304 | 0.4952 | 0.4636 |
| 0.304 | 0.4986 | 0.4460 |
| 0.300 | 0.4155 | 0.4266 |
| 0.296 | 0.4069 | 0.4076 |
| 0.296 | 0.4133 | 0.3883 |
| 0.292 | 0.3274 | 0.3687 |
| 0.292 | 0.3271 | 0.3488 |
| 0.288 | 0.3267 | 0.3286 |
| 0.288 | 0.3275 | 0.3081 |
| 0.284 | 0.3267 | 0.2873 |
| 0.284 | 0.2441 | 0.2663 |
| 0.280 | 0.2405 | 0.2450 |
| 0.280 | 0.2453 | 0.2234 |
| 0.280 | 0.1617 | 0.2018 |
| 0.276 | 0.1588 | 0.1799 |
| 0.276 | 0.1619 | 0.1578 |
| 0.276 | 0.1630 | 0.1355 |
|  |  |  |

Table 2.3-1 Measured versus theoretical dip angles for the shallowest reflection of the synthetic data example (Model 4)


Fig 2.3-3 A common source gather (a) and its instantaneous amplitude (b) and instantaneous phase (c) representations.
are made since the events are not as distinct as on the synthetic data set. Figure 2.3-3 displays an example of a common source gather and its instantaneous amplitude and instantaneous phase representations. Figure 2.3-4 is an example of the picks made by the complex attribute algorithm on real data. It is normal for the phase values to change rapidly from values near $+\pi$ to values near $-\pi$. This is commonly known as "phase wrap-around". One has to be very aware of phase wrapping when computing the instantaneous frequency.

Two of the major reasons for the increased number of mispicks made on real


Fig. 2.3-4 A real common source gather (a) and the picks made by the complex attribute picker (b), presented as small dipping bars. These bars are centred on the pick location and they dip with the measured apparent dip.
${ }^{6}$ Scheuer and Oldenburg (1988) preferred velocity filtering for their applications. Velocity filtering is avoided here because it totally removes steeply dipping energy that may provide some useful information. High cut filtering provides the opportunity for such energy to be preserved. Velocity filtering is also avoided as it requires the added computational expense of a 2D FFT whereas high cut filtering requires only 1D FFT's.
data are crossing events and noise. Both of these can make the frequency and wavenumber estimates unstable. These problems can be reduced by careful preliminary processing of the seismic data. Filtering in the $f-k$ domain can be used to remove linear, coherent noise and multiple reflections. Prediction error filtering in the $f-x$ domain has been used in all real examples to attenuate random noise. Finally, because maxima in instantaneous amplitude are being picked, it is imperative that the data be processed to produce a wavelet with approximately zero phase.

With the complex attribute picker it is possible for events to be aliased. That is, steeply dipping events may be picked with a dip opposite to the actual dip. As an example, see the picks made on the steeply dipping coherent energy of Fig 2.3-4. Aliasing can occur because the instantaneous wavenumber can be positive or negative (between $-\pi$ and $+\pi$ ). When the dip of an event becomes large enough so that its instantaneous wavenumber is slightly greater than $| \pm \pi|$, the sign and magnitude of the computed wavenumber will be wrong. The only way to avoid aliasing, apart from preliminary velocity filtering, is to high cut filter the input data. This reduces the rate of change of phase between adjacent traces and helps to keep the magnitude of the instantaneous wavenumber below $\pi$. Figure 2.3-5 has been high cut filtered at 35 Hz and shows reduced aliasing compared to Figure 2.3-4. However, high cut filtering also results in a decrease of resolution and tends to cause more problems with interfering events.

Scheuer and Oldenburg (1988) used a 2D FFT to compute the complex traces required for this type of data picking. It is more computationally intensive to use a 2 D FFT instead of a 1D FFT for each trace, and because only the negative temporal frequencies are zeroed (see earlier in this section), the results will be identical. Scheuer
data are crossing events and noise. Both of these can make the frequency and wavenumber estimates unstable. These problems can be reduced by careful preliminary processing of the seismic data. Filtering in the $f-k$ domain can be used to remove linear, coherent noise and multiple reflections. Prediction error filtering in the $f-x$ domain has been used in all real examples to attenuate random noise. Finally, because maxima in instantaneous amplitude are being picked, it is imperative that the data be processed to produce a wavelet with approximately zero phase.

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Scheuer and Oldenburg (1988) used a 2D FFT to compute the complex traces

[^8]${ }^{7}$ This simpler algorithm will actually pick the group velocity and not the phase velocity. This distinction is in principle immaterial for the marine data being considered in this thesis (since the near-surface is non-dispersive). However, it may be significant for seismic data acquired on land. In this case a correction to the dips may be required. Also, dip estimation based on amplitude may be more problematic in the presence of crossing events.


Fig. 2.3-5 The data used for Fig. 2.3-4 but high cut filtered to 35 Hz before picking. and Oldenburg made use of the $f-k$ domain for velocity filtering at the same time. All picks performed in this study used 1D FFT's.

The picking required here, for reflection tomography purposes, is only concerned with areas of maximum amplitude. Studying the rates of change of instantaneous phase may well be more detailed than is necessary. A much simpler approach would be to locate an amplitude maximum, search for the same maximum on adjacent traces, and compute the apparent dip. The promise of such an algorithm is further simplicity and efficiency. A program called NPCKSPC2 (see the appendix) was coded for the purpose and the flow of this algorithm is;
a) Read a common source gather of traces.


Fig. 2.3-5 The data used for Fig. 2.3-4 but high cut filtered to 35 Hz before picking.
and Oldenburg made use of the $f-k$ domain for velocity filtering at the same time. All picks performed in this study used 1D FFT's.

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${ }^{7}$ This simpler algorithm will actually pick the group velocity and not the phase velocity. This distinction is in principle immaterial for the marine data being considered in this thesis (since the near-surface is non-dispersive). However, it may be significant for seismic data acquired on land. In this case a correction to the dips may be required. Also, dip estimation based on amplitude may be more problematic in the presence of crossing events.
b) Compute the complex traces and the instantaneous amplitudes. The instantaneous amplitude trace defines the envelope of the input trace. When dealing with amplitudes, it is easier and simpler to use the amplitude of the trace envelope.
c) Search for amplitude maxima (not too close to each other).
d) Extract 12 instantaneous amplitude values around the maximum (see Figure 2.3-6).
e) Use quadratic interpolation to find the exact time of the maximum detected near sample 7. Given amplitude values, $A_{1}, A_{2}$ and $A_{3}$, with $A_{2}$ in the centre, the stationary point of the quadratic function passing through these is at

$$
t_{2}-\Delta t \frac{\left(A_{1}-A_{3}\right)}{2\left(A_{1}+A_{3}-2 A_{2}\right)}
$$

where $t_{2}$ is the travel time of the amplitude $A_{2}$. The stationary point is a maximum if $2 A_{2}>\left(A_{1}+A_{3}\right)$.
f) For the adjacent traces, use quadratic interpolation to find the maximum, but be aware that the maximum may have moved out of the bounds of the three samples. This is the reason for extracting five sample from the adjacent traces (see Figure 2.3-6).
g) Compute the two dips implied by the maxima times on


Fig. 2.3-6 Thirteen instantaneous amplitude values extracted around an amplitude maximum detected at sample 7.


Fig. 2.3-7 (a) a common source gather, (b) the picks made by the subroutine PICKSPC5 and, (c) the picks made by the simple amplitude picker (NPCKSPC2).
either side of the target trace. If these dips are very different, ignore the point.
h) Convert the dips to velocities with the known surface interval velocity and average.
i) Output velocity, along with source and receiver locations, to file for use by a similar receiver domain subroutine (NPCKCRP2).

An example of the picks made by this algorithm can be seen in Figure 2.3-7. The new picks displayed here appear more stable than those made by PICKSPC5 (Figure 2.37(b)).

### 2.4 Semi-analytical derivative estimates.

A key factor in the success of an inversion procedure is the accurate estimation of the sensitivity of the constraint statistic to changes of the model parameters. Here the constraint statistic is the sum of the squares of all the $X_{e r r}$ (see section 2.2) values,

$$
\begin{equation*}
C=\sum_{k} X_{e r r(k)}^{2} \tag{2.4-1}
\end{equation*}
$$

The model parameters are the discrete grid of interval velocities, $V_{i j}$. The required sensitivities are,

$$
\begin{equation*}
\frac{\partial C}{\partial V_{i j}}-\sum_{k} 2 X_{e r(k)} \frac{\partial X_{e r(k)}}{\partial V_{i j}}, \tag{2.4-2}
\end{equation*}
$$

where the derivatives on the right hand side are commonly called Frechet derivatives. These sensitivities define the local linear approximations of the $n$-dimensional C surface that are critical for the iterative inversion of the data (see section 4).

In both crosswell and reflection tomography, it is common to determine the Frechet derivatives analytically. Consider a raypath travelling through a velocity model defined by cells of constant velocity. The traveltime of this ray is,

$$
t=\sum_{k} l_{k} s_{k}
$$

where $l_{k}$ is the path length of the ray in the $k$ th cell and $s_{k}$ is the corresponding slowness ${ }^{1}$. Conventional tomography uses traveltime error as the constraint statistic,
${ }^{1}$ Slowness is the inverse of velocity, $s_{k}-\frac{1}{V_{k}}$.
and the Frechet derivatives are,

$$
\frac{\partial t}{\partial s_{k}}=l_{k}
$$

In this way, if slownesses are used as the model parameters, the Frechet derivatives are exactly the cell path lengths. However, this method assumes that the raypath does not change as the velocities change (ie. $\partial l_{k} / \partial s_{k}-0$ ). This assumption can be justified by invoking Fermat's principle which states that the raypath between any fixed points will be that with minimum traveltime. The raypath is stationary and will not change with small changes in velocity.

With the strategy of two-end raytracing without reflectors (see section 2.2), the reflecting surface is not defined at a fixed location and Fermat's principle does not apply. The reflection point is allowed to move arbitrarily and $\partial l_{k} / \partial s_{k}$ cannot be assumed to be zero. Analytic derivative estimates become very difficult, if not impossible. It is important that the Frechet derivatives are accurately estimated but computational efficiency is also paramount.

It is possible to continue with analytic derivative estimates if the range of velocity variations is restricted. Zhou et al., 1992, used analytic derivative estimates with a first-order surface fitted to the four nodes of a square cell. Biondi $(1988,1990)$ simplified the derivative estimation by allowing only velocity fields that can be defined by B-splines.

If analytic derivative estimates are not possible, finite-difference estimates can be used. This procedure requires the velocity at one of the nodes to be perturbed slightly, after which the ray is retraced and it is observed how the traveltime (or $X_{e r r}$ )
changes. This has to repeated for every velocity node and every ray. It quickly becomes very expensive in terms of computation time. Sword (1986) reduced the expense by performing fewer perturbations and propagating the results with a transfer matrix. With the triangular cells and arbitrary velocity variations of the current study, analytic derivative estimates are not possible, but accurate derivative estimates can be obtained with a modified finite-difference scheme.

Consider a single ray passing near a given velocity node, as in Figure 2.4-1. When considering the effect of a perturbation of this velocity node, it is recognised that the travel path in a six cell area of influence will be directly affected. An efficient method of estimating the sensitivity of $X_{e r r}$ with respect to velocity ( $\partial X_{e r r} / \partial V_{i j}$ ) will be


Fig. 2.4-1 A single raypath. If the velocity at A is perturbed, it will directly affect the travel path within a six cell zone of influence (shaded).
achieved if the raypath outside the area of influence is assumed to maintain its shape and is only re-oriented by the raypath changes within the area of influence. In this way the only extra ray tracing required will be within the six cell area of influence. The complete raypath will not have to be retraced as is necessary for finite-difference estimates.

Figure 2.4-2 demonstrates the principle of the proposed Frechet derivative estimation algorithm. The raypath shape from any cell to the end of the ray is assumed fixed, and is effectively defined by a vector $\Re$. The velocity perturbation of the node in question alters the raypath within the zone of influence, giving a new exit location and angle. These new exit parameters can be used to re-orient $\Re$, enabling a new $X_{\text {err }}$


Fig 2.4-2 The perturbed raypath within a cell results in a re-oriented vector $\Re$.
to be computed. Along with the original $X_{e r r}$, this value is used to estimate the required Frechet derivative. It is expected that these estimates will be accurate if the perturbation of the subject velocity node is small. A small perturbation also allows each cell of the area of influence to be handled separately with the results added together (as indicated in Figure 2.4-2). If the velocity perturbations are too large, the change in exit location and angle may be large enough to significantly change the shape of the rest of the raypath.

The code written to compute these semi-analytic ${ }^{2}$ derivative estimates is called SADE2 and can be found in the appendix. In order to make these estimates, SADE2 requires a number of values to be computed and placed in 'common blocks' for the raytracing subroutine XERRRT (also see the appendix). These values are;

1) upon exit of each cell - the cell number, the exit coordinates, the exit angle and the travel time at the moment of exiting,
2) the final depth (where the total travel time matches the data travel time) finalz,
3) the final x-coordinates of both the source and receiver ends of the ray finalx1 and finalx2,
4) the rates of change of $x$-coordinate with respect to depth at the end of each half of the ray $-d x d z 1$ and $d x d z 2$,
5) the rates of change of time with respect to depth at the end of each half of the ray $-d t d z 1$ and $d t d z 2$, and

[^9]6) the local velocity at the end of each half of the ray - finalv1 and finalv2.

All of this information is used by the SADE2 subroutine for the estimation of the derivative $\partial X_{e r r} / \partial V$, by re-orienting
the vector $\Re$ (see Figure 2.4-2) and computing the new depth at which the $X_{\text {err }}$ will be evaluated.


Fig. 2.4-3 The velocities of the model used to check the semi-analytic derivative estimates. The node spacing is $50 \times 50$ metres.

A synthetic data set was used to
verify the accuracy of these semi-analytic derivative estimates. The interval velocities of this model consist of a linear background increasing from $1500 \mathrm{~m} / \mathrm{s}$ to $3000 \mathrm{~m} / \mathrm{s}$, with a square anomaly of $2250 \mathrm{~m} / \mathrm{s}$ in the centre (see Figure 2.4-3). Rays were traced though this model with regular source and receiver locations and a range of surface angles (see Figure 2.4-4). The ray tracing resulted in 1312 valid rays (only 184 of these shown in Figure 2.4-4) within this model of $41 \times 41$ velocity nodes.

The most concise way to evaluate the semi-analytic derivative estimates is through the gradient of the constraint statistic, $\nabla C$. This gradient is calculated using the Frechet derivatives and is used to drive the inversion of the data (see equation (2.42)). The semi-analytic estimates are compared to fully ray traced finite-difference estimates in Figures 2.4-5 and 2.4-6. The finite-difference estimates were computed using a subroutine called FRECHX (see the appendix). Figures 2.4-5 and 2.4-6 are quite similar with one of the major differences being the magnitude of the derivative


Fig. 2.4-4 A subset of the rays traced through the model of Figure 2.4-3. 192 rays are displayed but only 184 of these were accepted as data.
estimates. The finite-difference estimates (Figure 2.4-5) are generally larger than their semi-analytic counterparts (Figure 2.4-6), especially for the shallower velocity nodes (small velocity node numbers). The semi-analytic estimates appear to be reasonable.

Both the finite-difference and semi-analytic estimates were used to complete an inversion of the synthetic data described above. The final images of these inversions are displayed in Figures 2.4-7 and 2.4-8. Surprisingly, the semi-analytic derivative estimates have resulted in a much improved inversion. The comparison of the inversion results can be found in Table 2.4-1. The semi-analytic derivative estimates have resulted in a smaller final $C$ value, in less iterations. It was expected that the semianalytic estimates would not be any better than the finite-difference ones; this test was only performed to see how close the semi-analytic estimates were to the finite difference estimates.

Fig. 2.4-5 Gradient estimates using FRECHX, the finite-difference derivative estimator.


Fig. 2.4-6 Gradient estimates using SADE2, the semi-analytic derivative estimator.


Fig. 2.4-7 The final image after an inversion with finite difference derivative estimates.


Fig. 2.4-8 The final image after an inversion with semi-analytic derivative estimates (SADE2).

| ITERATION | FINITE-DIFFERENCES <br> C Value / Expected C | SEMI-ANALYTIC <br> C Value / Expected C |
| :---: | :---: | :---: |
| 1 | $974515 / 809486$ | $974613 / 611467$ |
| 2 | $846018 / 684663$ | $611756 / 421692$ |
| 3 | $741426 / 601354$ | $421505 / 294016$ |
| 4 | $598125 / 477003$ | $292685 / 217063$ |
| 5 | $502037 / 428042$ | $206046 / 162009$ |
| 6 | $432897 / 376930$ | $159798 / 135241$ |
| 7 | $377288 / 275676$ | $130242 / 116015$ |
| 8 | $280804 / 251546$ | $117523 / 106245$ |
| 9 | $251746 / 214859$ | $103885 / 95554$ |
| 10 | $217855 / 206648$ | $97089 / 90108$ |
| 11 | $206733 / 197700$ | $89065 / 83378$ |
| 12 | $197790 / 191154$ | $84428 / 79377$ |
| 13 | $191685 / 188650$ | $78817 / 74132$ |
| 14 | $188519 / 180016$ | $75214 / 71153$ |
| 15 | $186743 / 178709$ | $71243 / 67595$ |
| 16 | $178885 / 165109$ | $68617 / 65491$ |
| 17 | $165724 / 161598$ | $65655 / 62735$ |
| 18 | $161781 / 156352$ | $63625 / 61045$ |
| 19 | $156854 / 154517$ | $61166 / 58806$ |
| 20 | $154527 / 145857$ | $59608 / 57480$ |
| 21 | $146436 / 143476$ | $57377 / 55394$ |
| 22 | $147074 / 144237$ | $56214 / 54403$ |
| 23 | $144391 / 140852$ | $54192 / 52495$ |
| 24 | $141116 / 139480$ | $53304 / 51730$ |
| 25 | $139531 / 136425$ | $51529 / 50060$ |
| 26 | $136874 / 135265$ | $50873 / 49499$ |
| 27 | $135396 / 133137$ |  |
| 28 | $133416 / 131460$ |  |
|  |  |  |

Table 2.4-1 The results of the inversion based on semi-analytic derivative estimated compared to those based on finite-difference estimates. The expected C value is that which the inversion predicts will be achieved in the next iteration.

A couple of features have been introduced to the semi-analytic derivative estimator (SADE2) that may affect its performance compared to the finite-difference estimator (FRECHX). The first feature was introduced to guard against problems that may arise if the vector $\Re$ (see Figure 2.4-2) has to be re-oriented too far from its original location. If the change in $X_{e r r}$ is more than 10 metres, a zero is returned for the derivative involving that particular ray and velocity node. The need for this was rare with the current model, however, ignoring unstable points may have helped stabilise the inversion. Another less important feature has been added to account for the rare possibility of the new exit location of the ray actually being deeper than the required depth for the evaluation of $X_{e r r}$. Once again, these occurrences were ignored by returning a zero derivative.

One last problem for the semi-analytic derivative estimator is the possibility of the new exit point from the cell being a long way from the original exit point. Figure 2.4-9 demonstrates one set of circumstances where this could happen. In this case, the ray will now pass through a cell not traversed previously. It is difficult to maintain the assumption that the rest of the raypath can be defined by the vector $\Re$. Once again, this occurrence would be rare but it might be better not to attempt estimates


Fig. 2.4-9 The new exit point, B, can be a long way from the original exit point, A. Even to the point of entering a cell not previously traversed.
in these cases.
The motivation for developing the semi-analytic derivative estimator was improved computational efficiency. The intuitive ideas underlying its implementation seem to have worked satisfactorily in practice. The 1312 ray example described earlier used 1700 seconds of CPU time ${ }^{3}$ per iteration with finite-difference approximations but only 75 seconds for the semi-analytic estimates. A twenty fold reduction in run time has been achieved. It is interesting to note that of the 75 seconds per iteration, 65 seconds were taken up by ray tracing and derivative estimation - only 10 seconds for the inversion. Further efficiency improvements must come from the ray tracing software. This is a rather small model though; the proportional expense changes dramatically for larger models.

An interesting technique that may also be useful for the current problem is "approximate inverse mapping" (Oldenburg and Ellis, 1991). This uses a relatively simple approximate inverse mapping from the data space to the model space and determines correction vectors to account for the approximation. Linearisation is not necessarily required. This type of algorithm has not been tested within the current work.

[^10]2.5 Stages of decreasing smoothing.

Once the traveltime data has been obtained, an initial velocity model is required for the commencement of the iterative inversion of these data. There is rarely full confidence in the initial velocity model. Usually some aspects of the model cause concern, for example, the best velocity for a particular layer that is assumed to be homogeneous. With such concerns, it seems inconsistent for the inversion process to consider individual changes to each of the closely spaced, discrete velocity values. It would seem more appropriate to change the velocities of complete geologic units initially, and only later allow for higher spatial frequency updates. This would be tantamount to using the data to improve the starting model before allowing the standard inversion to begin.

Consider a synthetic velocity model consisting of an 11 x11 grid of velocity values. The velocities assigned to this grid are $2000 \mathrm{~m} / \mathrm{s}$ except for the $3 \times 3$ block of velocities in the centre which are $2050 \mathrm{~m} / \mathrm{s}$. That is, the model has a fast square shaped anomaly in the centre. A total of 192 raypaths were traced though this model to generate the traveltime data to be inverted. These raypaths were generated to symmetrically span the horizontal extent of the model with a uniform range of short to long offsets, along with a uniform range of surface angles to ensure good coverage at all depths. An initial constant velocity model of $2000 \mathrm{~m} / \mathrm{s}$ was used to begin the inversion that resulted in the image of Figure 2.5-1. This inversion did not successfully uncover the anomaly.

A potential problem with an inversion as described above is that the initially

| 2021 | 2000 | 1991 | 1998 | 2003 | 2010 | 2008 | 2003 | 1993 | 1998 | 2010 |
| :--- | :--- | :--- | :--- | :--- | :--- | :--- | :--- | :--- | :--- | :--- |
| 2009 | 2005 | 1989 | 2004 | 2007 | 2008 | 2011 | 2002 | 1988 | 2007 | 2006 |
| 2002 | 2023 | 1984 | 1992 | 2012 | 2008 | 2024 | 1997 | 1991 | 2006 | 2002 |
| 2000 | 2014 | 2003 | 1976 | 2007 | 2008 | 2029 | 1992 | 1997 | 2008 | 2000 |
| 2000 | 2005 | 2020 | 1956 | 2024 | 2000 | 2016 | 1977 | 2008 | 2003 | 2000 |
| 2000 | 2003 | 2022 | 1991 | 1988 | 1989 | 1981 | 1987 | 2012 | 2001 | 2000 |
| 2000 | 2001 | 2013 | 2012 | 1956 | 1979 | 1963 | 1993 | 2009 | 2000 | 2000 |
| 2000 | 2001 | 2007 | 2022 | 1962 | 1960 | 1954 | 1987 | 2005 | 2000 | 2000 |
| 2000 | 2000 | 2003 | 2009 | 1985 | 1944 | 1958 | 1992 | 2003 | 2000 | 2000 |
| 2000 | 2000 | 2000 | 2000 | 1995 | 1969 | 1984 | 1998 | 2000 | 2000 | 2000 |
| 2000 | 2000 | 2000 | 2000 | 2000 | 1996 | 1999 | 2000 | 2000 | 2000 | 2000 |

Fig. 2.5-1 An inversion of the 11x11 synthetic model data. The initial model was $2000 \mathrm{~m} / \mathrm{s}$ for every grid node. The values within the rectangle should be $2050 \mathrm{~m} / \mathrm{s}$.
(See the footnote on page 15)
traced raypaths may, in some cases, be far from the actual raypaths. This is because the anomaly is not defined by the initial model. Figure 2.5-2 demonstrates how the traced and actual ray locations can differ appreciably. It makes little sense to attempt complete correction of the velocity model based on such incorrect ray locations. An effort must be made to improve the ray locations before the final solution can be derived. As mentioned earlier in this section, the velocities of whole geologic units could be changed until the best simple velocity model is found. Such a new simple model should position the rays closer to their actual location and the high-fidelity inversion can be allowed to proceed with more confidence. Effectively, the traveltime data are used to improve the initial velocity model. However, as with the current synthetic example, the existence of geologic units (in this case, a square region of fast velocity) is not always known. It is desirable to find a procedure that will help improve the ray locations and is independent of the velocity field.

The standard inversion will consider changes to each discrete velocity value independently. Intuitively, high spatial frequency velocity updates will be of little use if


Fig. 2.5-2 A schematic demonstration of how different the actual and traced raypaths can be when the anomaly is not defined in the initial velocity model.
the rays are not near their correct locations. High spatial frequency changes will have little effect on the location of the rays and it will be difficult for the iterative inversion to approach the correct solution with such changes. Low spatial frequency, or smooth, velocity changes will have more effect on the locations of the rays. One can imagine an algorithm that allows only smooth velocity changes at first, effectively locating the raypaths, and only then allows higher frequency changes.

One such algorithm consists of blocking the velocity values for the early iterations (this type of procedure has been described by Williamson, 1990). Consider the $11 \times 11$ synthetic model described earlier. If it is extended by an extra row and column to a discrete $12 \times 12$ model, then $4 \times 4$ groups of velocity nodes can be "blocked" together for the early iterations of the inversion. Each of the sixteen velocities within a block will be changed equally. The sensitivities will be computed for velocity changes
of the block as a whole. The inversion initially becomes a $3 \times 3$ problem in which each cell consists of 16 sub-cells. The initial model will still be a uniform velocity of 2000 $\mathrm{m} / \mathrm{s}$, but the early blocked iterations effectively find a better low frequency model that will define the ray locations more accurately.

After nine iterations of such a blocked procedure, the image of Figure 2.5-3 results. Notice that each of the $4 \times 4$ blocks have the same velocity. The block containing the anomaly now has a velocity of $2037 \mathrm{~m} / \mathrm{s}$ which is approaching the true velocity of the anomaly, $2050 \mathrm{~m} / \mathrm{s}$. Blocking has produced a noticeable improvement over the unblocked inversion (Figure 2.5-1) even for this simple model. The raypaths generated in the blocked inversion should be better located. A further eight iterations were performed without any blocking (but with the model shown in Figure 2.5-3 as the starting model), the results of which can be seen in Figure $2.5-4$. The velocity values within the marked anomalous zone are close to the required $2050 \mathrm{~m} / \mathrm{s}$ but the residual effect of the blocked zone (dashed line) is still apparent. This example has shown that it can be very beneficial to attempt to correctly locate the rays before proceeding with the inversion. However, there is also evidence that the blocking algorithm may not be the best approach.

Williamson (1990) also explored this approach of blocking but referred to it as a variable parameterisation scale-length technique. His original motivation was to force the inverse problem to be temporarily over-parameterised, and hence stable. After noticing similar residual velocity problems as those demonstrated by Figure 2.5-4, Williamson developed a translation invariant approach with the same benefits as the blocking approach. He introduced a model covariance matrix to effect smoothing of the constraint statistic, $\boldsymbol{C}$. The smoothed constraint statistic for the current problem can be

| 1986 | 1986 | 1986 | 1986 | 1977 | 1977 | 1977 | 1977 | 1972 | 1972 | 1972 | 1972 |
| :--- | :--- | :--- | :--- | :--- | :--- | :--- | :--- | :--- | :--- | :--- | :--- |
| 1986 | 1986 | 1986 | 1986 | 1977 | 1977 | 1977 | 1977 | 1972 | 19772 | 1972 | 1972 |
| 1986 | 1986 | 1986 | 1986 | 1977 | 1977 | 1977 | 1977 | 1972 | 1972 | 1972 | 1972 |
| 1986 | 1986 | 1986 | 1986 | 1977 | 1977 | 1977 | 1977 | 1972 | 1972 | 1972 | 1972 |
| 1986 | 1986 | 1986 | 1986 | 2037 | 2037 | 2037 | 2037 | 2002 | 2002 | 2002 | 2002 |
| 1986 | 1986 | 1986 | 1986 | 2037 | 2037 | 2037 | 2037 | 2002 | 2002 | 2002 | 2002 |
| 1986 | 1986 | 1986 | 1986 | 2037 | 2037 | 2037 | 2037 | 2002 | 2002 | 2002 | 2002 |
| 1986 | 1986 | 1986 | 1986 | 2037 | 2037 | 2037 | 2037 | 2002 | 2002 | 2002 | 2002 |
| 2000 | 2000 | 2000 | 2000 | 1987 | 1987 | 1987 | 1987 | 2002 | 2002 | 2002 | 2002 |
| 2000 | 2000 | 2000 | 2000 | 1987 | 1987 | 1987 | 1987 | 2002 | 2002 | 2002 | 2002 |
| 2000 | 2000 | 2000 | 2000 | 1987 | 1987 | 1987 | 1987 | 2002 | 2002 | 2002 | 2002 |
| 2000 | 2000 | 2000 | 2000 | 1987 | 1987 | 1987 | 1987 | 2002 | 2002 | 2002 | 2002 |

Fig. 2.5-3 After 9 iterations of the "blocked" inversion procedure, the anomaly is better defined. The anomalous zone is marked by the solid line.


Fig. 2.5-4 A further eight iterations with no blocking does not remove the effect of the earlier blocked iterations. The anomalous zone is marked with a solid line. The blocked zone is marked with a dashed line.
expressed as,

$$
\begin{equation*}
C-\mathbf{X}_{\mathrm{err}}^{\mathrm{T}} \mathrm{C}_{\mathrm{d}}^{-1} \mathbf{X}_{\mathrm{err}}+\left(\mathbf{v}-\mathbf{v}_{0}\right)^{\mathbf{T}} \mathrm{C}_{\mathrm{m}}^{-1}\left(\mathbf{v}-\mathbf{v}_{0}\right) \tag{2.5-1}
\end{equation*}
$$

where $\mathbf{X}_{\text {err }}$ is the column vector of $X_{e r r(i)}$ values for each ray, $\mathbf{C}_{\mathbf{d}}$ is the data covariance matrix (currently assumed to be the identity matrix), $\mathbf{v}$ are the new velocities derived by this iteration, $\mathbf{v}_{0}$ are the current velocities and $\mathbf{C}_{\mathrm{m}}$ is the model covariance matrix. Williamson (1990) added two-dimensional Gaussian shaped functions to the model covariance matrix, effectively forcing the velocity updates made by each iteration to be as smooth as these functions. This approach treats every velocity value equally. The velocities are not grouped arbitrarily. Smooth changes will be enforced with limited side-effects.

The new constraint statistic of equation (2.5-1) was adopted in the inversion process described in this thesis. The first and second derivatives of $C$ will be needed for this purpose. The first derivative is given by the gradient vector,

$$
\begin{equation*}
\nabla C-2 \nabla \mathbf{X}_{\mathrm{err}}^{\mathrm{T}} \mathbf{X}_{\mathrm{err}}+2 \mathrm{C}_{\mathrm{m}}^{-1}\left(\mathbf{v}-\mathbf{v}_{0}\right), \tag{2.5-2}
\end{equation*}
$$

and the corresponding second derivative is,

$$
\begin{equation*}
\nabla \nabla C-2 \nabla \mathbf{X}_{\mathrm{err}}^{\mathbf{T}} \nabla \mathbf{X}_{\mathrm{err}}+2 \nabla \nabla \mathbf{X}_{\mathrm{err}}^{\mathbf{T}} \mathbf{X}_{\mathrm{err}}+2 \mathrm{C}_{\mathrm{m}}^{-1} \tag{2.5-3}
\end{equation*}
$$

Each of equations (2.5-1), $(2.5-2)$ and $(2.5-3)$ are identical to the conventional expressions except for their final terms. In each case, the last term suggests that it may be necessary to calculate the inverse of $\mathrm{C}_{\mathrm{m}}$. However, this computational expense is avoided since the above vectors are in a space dual to the model space. The dual space is the space of all linear forms of the original space (see Tarantola, 1987, section
4.4.2). Here, the above vectors are in the dual space by the action of $C_{m}$. To obtain the vector locations in the model space, the above forms must be operated on by $\mathrm{C}_{\mathrm{m}}$ (see Tarantola, 1987, equation (4-68)). Therefore,

$$
\begin{equation*}
\nabla C^{\prime}=2 C_{m} \nabla X_{e r r}^{T} X_{e r r}+2\left(v-v_{0}\right) \tag{2.5-4}
\end{equation*}
$$

and,

$$
\begin{equation*}
\nabla \nabla C^{\prime}=2 C_{m} \nabla X_{e r r}^{T} \nabla X_{e r r} C_{m}+2 C_{m} \nabla \nabla X_{e r r}^{T} X_{e r r} C_{m}+2 C_{m} \tag{2.5-5}
\end{equation*}
$$

(the symmetry of the covariance matrices has been used here). The second term of equation (2.5-5) is usually ignored (see Tarantola, 1987 (page 194), and Kennett et al., 1988) because $\nabla V X_{\text {err }}$ diminishes as the data fit improves and not least because it is also difficult to compute. The actual equation used is then,

$$
\begin{equation*}
\nabla \nabla C^{\prime}=2 \mathrm{C}_{\mathrm{m}} \nabla \mathrm{X}_{\mathrm{err}}^{\mathrm{T}} \nabla \mathrm{X}_{\mathrm{err}} \mathrm{C}_{\mathrm{m}}+2 \mathrm{C}_{\mathrm{m}} \tag{2.5-6}
\end{equation*}
$$

The Gaussian functions used by Williamson are characterised by a half-width, $L$. The value of the function at a distance $d$ from the subject point is then,

$$
\begin{equation*}
g=\exp \left[-\frac{1}{2}\left(\frac{d}{L}\right)^{2}\right] \tag{2.5-7}
\end{equation*}
$$

This function is calculated by the subroutine GAUSCAL3 (see the appendix) and the results are stored in a compressed format before the commencement of the inversion. Most of the work involved with smoothing application is controlled by the group of subroutines that perform the inversion and which are driven by the subroutine MEMNL\$8 (see the appendix) ${ }^{1}$.

[^11]Instead of arbitrarily blocking groups of velocities, Williamson (1990) suggested that the early iterations be performed with Gaussian smoothers that have a half-width of the order of the dimensions of the model. Once these iterations make no further $C$ reductions, Williamson reduced the half-width by a factor of two and allowed the next stage of the inversion to proceed. This approach is continued until no smoothing at all is applied. The result is to initially extract only the very low frequency


Fig. 2.5-5 The interval velocities and dimensions of the model used to create the synthetic data. features of the model and then slowly extract higher and higher frequency features. The effect of this algorithm is to impose a bias towards positioning of the raypaths early in the inversion and then gradually allowing the details of the velocity anomalies to be extracted.

It is useful to evaluate this technique on some synthetic data. The interval velocities and dimensions of the model to be used are shown in Figure 2.5-5. A random number generator was used to randomly select source and receiver locations as well as source and receiver surface angles, for a total of 800 rays. These rays were traced through the model to derive the traveltime data ${ }^{2}$. The velocities were defined on

[^12]a $50 \times 50$ metre grid. A starting velocity model was prepared with the correct velocity for the first layer (ie. $2000 \mathrm{~m} / \mathrm{s}$ ) and with a linear velocity increase of $0.9 \mathrm{~m} / \mathrm{s}$ per metre from the base of this layer (see Figure 2.5-6). This starting model implies almost total ignorance of the model. In general, if no other information is at hand, a simple linear increase with depth would be used as the starting model.

An inversion of these data was performed using the stages of decreasing smoothing approach described earlier. Figures 2.5-7 through 2.5-12 display the inverted image after each of the stages of smoothing. The respective Gaussian halfwidths are $1600,800,400,200$ and 100 metres with the final stage having no smoothing. The features of the model are gradually extracted, with the final image (Figure 2.5-12) displaying a good expression of the anomaly ${ }^{3}$, especially considering the poor starting velocity model. Fig $2.5-13$ shows a graph of the decrease of the constraint statistic during the 201 iterations required to achieve the final result. The effect of the stages of decreasing smoothing is clear. The smoothing width is reduced when the value of $C$ stops decreasing. This gives Figure 2.5-13 its unusual step-like appearance.

The same data were also inverted without any stages of decreasing smoothing. Figure 2.5-14 shows the final image of this inversion. It is difficult to see an expression of the velocity anomaly in these results. A graph of the change in constraint statistic, $C$, is shown in Figure 2.5-15. Notice that the final $C$ value is almost double that of the inversion with stages of smoothing (Fig. 2.5-13). The inversion without smoothing has undoubtedly encountered a local minimum of $C$ and converged onto

[^13]

Fig. 2.5-6 The starting velocity model for the inversion of the synthetic data derived from the model of Figure 2.5-5


Fig. 2.5-8 The image after the stage of the inversion that used Gaussian smoothers with 800 metre half-widths.

Fig. 2.5-7 The image after the stage of the inversion that used Gaussian smoothers with 1600 metre half-widths.


Fig. 2.5-9 The image after the stage of the inversion that used Gaussian smoothers with 400 metre half-widths.


Fig. 2.5-10 The image after the stage of the inversion that used Gaussian smoothers with 200 metre half-widths.


Fig. 2.5-12 The image after the final stage of the inversion that free from smoothing.


Fig. 2.5-11 The image after the stage of the inversion that used Gaussian smoothers with 100 metre half-widths.


Fig. 2.5-13 The values of the constraint statistic. The stepped appearance is caused by the stages of decreasing smoothing.


Fig. 2.5-14 The final image from an inversion without smoothing stages.


Fig. 2.5-15 The change in constraint statistic values for the inversion without stages of smoothing.
that solution.
The inversion of the traveltime data consists of finding the velocity model that reduces $C$ to its lowest possible value. This is complicated because $C$ is an $n$-dimensional ${ }^{4}$ non-linear function. The most efficient way to search for the minimum is to make local linear (or quadratic) approximations to this non-linear function, and iteratively make small steps of decreasing $C$ values. The problem with following a path of descent is that local minima of $C$ can trap the inversion and prevent it from finding the global minimum (schematically shown in Figure 2.5-16).

When stages of smoothing are used in the inversion, the constraint statistic, $C$-surface is smoothed during the early stages (recall equation $2.5-1$ ). This has the

[^14]potential to smooth out local minima temporarily (Figure $2.5-17$ ) and then, as the smoothing decreases, focus on the global minimum. This schematical representation effective can also be used to demonstrate circumstances where the s smoothing of $C$ may not be $^{C}$ successful. Consider Figures $2.5-18$ and $2.5-19$. In the first case a residual local minimum remains after smoothing and the inversion is still trapped. Heavier smoothing would have been more appropriate. In the second case (Figure 2.5-19) the inversion actually moves away from the global minimum - less smoothing may reverse this trend. These examples demonstrate that ideal smoothing is data dependent and difficult to define. With the assumption that the constraint surface is reasonably stable near the global minimum and that the starting model is not too far from the actual model, then the technique of stages of decreasing smoothing, as described by Williamson (1990) and discussed above, will be generally successful.

Jupp and Vozoff's (1975) paper provides the theoretical justification of the technique of stages of decreasing smoothing. A short review of their work is useful here.

In terms of the general inversion problem, let $\mathbf{x}$ represent the model parameters, $\mathbf{d}$ the data values and $\mathbf{g}(\mathbf{x})$ the modelled data values. The inversion procedure attempts to minimise,

$$
(\mathbf{d}-\mathbf{g}(\mathbf{x}))^{2} .
$$

This is the general constraint statistic. The truncated Taylor series expansion gives,

$$
g(x+\delta x)-g(x)+J \delta x
$$

where, $J_{i j}=\frac{\partial g_{i}}{\partial x_{j}}$ are the Frechet derivatives.


Fig. 2.5-16 Local minima (LM) can prevent the inversion path from reaching the global minimum (GM). SM is the location of the starting model in this space.


Fig. 2.5-17 A schematic cross-section of the constraint statistic surface. Heavy smoothing in early stages can smooth out local minima (LM) while later stages allow focussing on the global minimum (GM).


Fig. 2.5-18 The constraint surface (solid line) is smoothed (dashed line) but the starting model (SM) is still trapped be a residual local minimum at point A . Heavier smoothing is required.


Fig. 2.5-19 The inversion moves away from the global minimum (GM) even after smoothing (dashed line). Less smoothing may be more appropriate.

This equation can also be expressed as,

$$
\delta \mathbf{g} \sim \mathbf{J} \delta \mathbf{x}
$$

Using the singular value decomposition (SVD) of J, Jupp and Vozoff (1975) show that the components of the model that have small singular values have very little effect on $\delta \mathrm{g}$. Changes of these components will have little effect on the constraint statistic.

Such components of the model are unstable.
The simplest way to avoid problems with small singular value components of the model is to specify a singular value threshold. Changes of components with singular values below the threshold are not considered. Jupp and Vozoff define this as the truncation strategy. Rather than employ such an abrupt cutoff, Jupp and Vozoff also considered a more gentle 'damping' of the small singular value components. They claim that the Marquardt method (1963) is one such damping technique but generalise this to a parametric damping function that encompasses both the Truncation and Marquardt approaches, along with others.

Near the end of their paper, Jupp and Vozoff actually suggest a practical strategy of raising the threshold for the beginning of the inversion "so that only the basic features of the model will be resolved". Reducing this threshold as the inversion proceeds seems to be closely related to Williamson's stages of decreasing smoothing approach. As Jupp and Vozoff suggested, if the components with large singular values are the smoother, or more basic, components, then the SVD and stages of smoothing techniques are the same in principle. It can be concluded that the stages of smoothing strategy is effectively damping the unstable components of the model.

Marquardt's (1963) paper contains an excellent description of the ideas that led
him to develop his damped inversion. He did not use singular values, but Jupp and Vozoff (1975) proved that his method is equivalent to a particular type of damping applied to the singular values. Marquardt noticed that the method of steepest descent made very good progress at the beginning of an inversion but converged slowly at the end, whereas the Gauss-Newton method may be slow at the start but converges rapidly at the end. In an effort to try and get the benefits of both procedures, Marquardt added a constant to the diagonal of the Frechet derivative matrix that resulted from the truncated Taylor series expansion (see above),

$$
\delta g=(\mathbf{J}+\lambda \mathbf{I}) \delta x
$$

and showed that with $\lambda=\infty$ one get the steepest descent algorithm and with $\lambda=0$, the Gauss-Newton algorithm. By varying $\lambda$ as the inversion proceeds, one can change gradually from the steepest descent algorithm to the Gauss-Newton algorithm. Jupp and Vozoff (1975) showed that this is equivalent to their damped singular value approach.
2.6 Weighted least-squares.

When the surface angles are picked from real seismic data (see section 2.3) some mispicks are unavoidable. These mispicks can result in some very large $X_{\text {err }}$ values even with an accurate velocity model. Even without mispicking, some rays can be very sensitive and have large $X_{e r r}$ 's for quite accurate models. Since the constraint statistic squares the $X_{e r r}$ values, a few such large $X_{e r r}$ 's can stifle the inversion process. In general, rays with smaller $X_{\text {err }}$ 's are more likely to be reliable. It is reasonable to apply a weighting to the data to bias the inversion toward rays with smaller $X_{e r r}$ 's.

Recall the expression for the constraint statistic (equation 2.5-1),

$$
\begin{equation*}
C=\mathbf{X}_{\mathrm{err}}^{\mathbf{T}} \mathrm{C}_{\mathrm{d}}^{-1} \mathbf{X}_{\mathrm{err}} \tag{2.6-1}
\end{equation*}
$$

but here ignoring the additive smoothing term. If the data covariance matrix is diagonal, this function can be expressed in suffix notation as,

$$
\begin{equation*}
C=\sum_{i} \frac{X_{e r(i)}^{2}}{\sigma_{i}^{2}}, \tag{2.6-2}
\end{equation*}
$$

where $\sigma_{i}^{2}$ is the variance associated with the $i$ th data value. From the above discussion it is reasonable to assume that there is a relationship between $\sigma_{i}^{2}$ and the magnitude of $X_{e r(i)}$. For the purposes of the current study, it is simpler to assign weights, $w_{i}$, instead of variances $\sigma_{i}^{2}$, such that,

$$
\begin{equation*}
C-\sum_{i} X_{e r(i)}^{2} w_{i} \tag{2.6-3}
\end{equation*}
$$

with $w_{i}=1 / \sigma_{i}^{2}$. These weights are to be chosen such that the inversion process is biased toward rays with smaller $X_{e r r}$ 's.

After experimentation, the following modified Hanning function was used as a satisfactory specification for the weights,

$$
\begin{align*}
w_{i} & =\left[\frac{3}{4}+\frac{1}{4} \cos \left(\frac{\pi X_{e r r}(i)}{X_{r e f}}\right)\right]^{2} & & \text { for } X_{e r r(i)}<X_{r e f}  \tag{2.6-4}\\
& =0.25 & & \text { for } X_{e r r(i)} \geq X_{r e f}
\end{align*}
$$

The parameter $X_{r e f}$ is used to control the extent of the weighting. The value of $X_{r e f}$ was originally set to somewhat arbitrary 300 metres.

The incorporation of weights into the inversion process initially seemed straightforward. However, with the weights as defined in equation (2.6-4), there will be changes in the weights between iterations. Changing the velocity model changes the $X_{e r r(i)}$ 's and hence also changes the $w_{i}$ 's. This means that $\nabla C$ and $\nabla \nabla C$ will have extra terms due to the weights.

Consider the gradient of the constraint statistic,

$$
\begin{equation*}
\frac{\partial C}{\partial v_{j}}-\sum_{i}\left[2 X_{e r(i)} \frac{\partial X_{e r r(i)}}{\partial v_{j}} w_{i}+X_{e r(i)}^{2} \frac{\partial w_{i}}{\partial v_{j}}\right] . \tag{2.6-5}
\end{equation*}
$$

The second term of this expression would not normally be present since $\frac{\partial w_{i}}{\partial v_{j}}$ is usually
zero. The second derivative is,

$$
\begin{align*}
\frac{\partial C}{\partial v_{k} \partial v_{j}}-\sum_{i} & {\left[2 \frac{\partial X_{e r(i)}}{\partial v_{k}} \frac{\partial X_{e r r(i)}}{\partial v_{j}} w_{i}+2 X_{e r(i)} \frac{\partial X_{e r r(i)}}{\partial v_{j}} \frac{\partial w_{i}}{\partial v_{k}}\right.}  \tag{2.6-6}\\
& \left.+2 X_{e r(i)} \frac{\partial X_{e r r(i)}}{\partial v_{k}} \frac{\partial w_{i}}{\partial v_{j}}+X_{e r(i)}^{2} \frac{\partial^{2} w_{i}}{\partial v_{k} \partial v_{j}}\right],
\end{align*}
$$

where the term containing $\frac{\partial^{2} X_{e r(i)}}{\partial v_{k} \partial v_{j}}$ has once again been ignored (see section 2.5). The complexity of these equations for the gradients of $C$ suggest that applying weights in the inversion process may be difficult. The second derivative has three extra terms.

The derivative of the weight vector must be evaluated. The first derivative is,

$$
\begin{align*}
\frac{\partial w_{i}}{\partial v_{j}} & -2\left[\frac{3}{4}+\frac{1}{4} \cos \left(\frac{\pi X_{e r r(i)}}{X_{r e f}}\right)\right]\left\{-\frac{1}{4} \sin \left(\frac{\pi X_{e r(i)}}{X_{r e f}}\right)\right\}-\frac{\pi}{X_{r e f}} \frac{\partial X_{e r(1)}}{\partial v_{j}}  \tag{2.6-7}\\
& =a_{i} \frac{\partial X_{e r r(i)}}{\partial v_{j}}
\end{align*}
$$

where

$$
\begin{align*}
a_{i} & =\frac{-\pi}{2 X_{r e f}} \sin \left(\frac{\pi X_{e r r(i)}}{X_{r e f}}\right)\left(\frac{3}{4}+\frac{1}{4} \cos \left(\frac{\pi X_{e r r(i)}}{X_{r e f}}\right)\right) \\
& =\frac{-\pi}{2 X_{r e f}} \sin \gamma\left(\frac{3}{4}+\frac{1}{4} \cos \gamma\right), \tag{2.6-8}
\end{align*}
$$

and $\gamma-\frac{\pi X_{\text {err(i) }}}{X_{r e f}}$. The second derivative is,

$$
\begin{aligned}
\frac{\partial^{2} w_{i}}{\partial v_{k} \partial v_{j}}= & \frac{\partial}{\partial X_{e r r(i)}}\left\{\frac{\partial w_{i}}{\partial v_{j}}\right\} \frac{\partial X_{e r r(i)}}{\partial v_{k}} \\
= & \left\{\frac{-\pi}{2 X_{r e f}} \cos \gamma \frac{\pi}{X_{r e f}}\left(\frac{3}{4}+\frac{1}{4} \cos \gamma\right) \frac{\partial X_{e r(i)}}{\partial v_{j}}\right. \\
& -\frac{\pi}{2 X_{r e f}} \sin \gamma\left[-\frac{1}{4} \sin \gamma \frac{\pi}{X_{r e f}} \frac{\partial X_{e r r(i)}}{\partial v_{j}}\right. \\
& \left.\left.+\left(\frac{3}{4}+\frac{1}{4} \cos \gamma\right) \frac{\partial^{2} X_{e r r(i)}}{\partial v_{k} \partial v_{j}} \frac{\partial v_{k}}{\partial X_{e r r(i)}}\right]\right\} \frac{\partial X_{e r r(i)}}{\partial v_{k}}
\end{aligned}
$$

or, on simplification,

$$
\begin{aligned}
\frac{\partial^{2} w_{i}}{\partial v_{k} \partial v_{j}} & \left.-\left\{\frac{-\pi^{2}}{2 X_{r e f}^{2}}\left(\frac{3}{4} \cos \gamma+\frac{1}{4} \cos ^{2} \gamma-\frac{1}{4} \sin ^{2} \gamma\right)\right\}\right\} \frac{\partial X_{e r r(i)}}{\partial v_{k}} \frac{\partial X_{e r r(i)}}{\partial v_{j}} \\
& =b_{i} \frac{\partial X_{e r r(i)}}{\partial v_{k}} \frac{\partial X_{e r(i)}}{\partial v_{j}},
\end{aligned}
$$

where the term containing $\frac{\partial^{2} X_{e r r(i)}}{\partial v_{k} \partial v_{j}}$ has been ignored. Using equations (2.6-7) and
(2.6-9) the derivatives of the constraint statistic become,

$$
\begin{equation*}
\frac{\partial C}{\partial v_{j}}-\sum_{i} \frac{\partial X_{e r r(i)}}{\partial v_{j}}\left[2 w_{i} X_{e r r(i)}+a_{i} X_{e r r(i)}^{2}\right] \tag{2.6-10}
\end{equation*}
$$

and,

$$
\begin{equation*}
\frac{\partial^{2} C}{\partial v_{k} \partial v_{j}}-\sum_{i} \frac{\partial X_{e r r(i)}}{\partial v_{k}} \frac{\partial X_{e r r(i)}}{\partial v_{j}}\left[2 w_{i}+4 a_{i} X_{e r r(i)}+b_{i} X_{e r r(i)}^{2}\right] \tag{2.6-11}
\end{equation*}
$$

which shows that applying the weights may not be as difficult as first suggested by equations (2.6-5) and (2.6-6). The variables $a_{i}$ and $b_{i}$ are both simple functions of
$X_{e r r(i)}$, as is $w_{i}$. A subroutine is called before the execution of the inversion process to compute the values of $w_{i}, a_{i}$ and $b_{i}$ for the current iteration. This subroutine is called VARIAN and the code is listed in the appendix.

It is crucial that the weights do not decrease too rapidly with increasing $X_{e r r(i)}$. If this happens it is possible that increasing $X_{\text {err(i) }}$ could result in a reduction of the constraint statistic, $C$ (see equation (2.6-3)). Such a condition would prevent the successful inversion of the data. It is necessary that,

$$
\begin{equation*}
\frac{\partial C}{\partial X_{e r r(i)}}-2 X_{e r r(i)} w_{i}+X_{e r r(i)}^{2} \frac{\partial w_{i}}{\partial X_{e r r(i)}} \tag{2.6-12}
\end{equation*}
$$

be positive for all $X_{e r r(i)}$. Therefore,

$$
\begin{equation*}
\frac{\partial w_{i}}{\partial X_{e r r(i)}}>\frac{-2 w_{i}}{X_{e r r(i)}} \tag{2.6-13}
\end{equation*}
$$

must be satisfied to prevent the weights from decreasing too rapidly with increasing $X_{e r r(i)}$.

A direct comparison of inversion results without and with weights can be seen in Figures 2.6-1 and 2.6-2. The synthetic data used were that same as described in section 2.5 (see Figures $2.5-5$ and $2.5-6$ ). It is clear that the inversion with $X_{e r r(i)}$ weighting has achieved superior results.


Fig. 2.6-1 The inversion of data described in section 2.5. No weighting applied.


Fig. 2.6-2 The inversion of the same data as Figure 2.6-1. This time with weighting applied (as in equation 2.6-4).

## §3 THE ENTROPY PRINCIPLE - THEORY

### 3.1 Introduction

Tomographic image reconstruction in geophysical applications is generally an underdetermined and somewhat ill-posed problem. The data are almost always incomplete and noisy, and it is found that many different images can represent satisfactory solutions. Since typical images consist of tens of thousands of elements, it is too difficult to display an analysis of error bounds, and the geophysicist is compelled to select a single image as the solution. This is not a very unusual situation as scientific data is often given as the line of best fit through the given data values, even though there are many such lines which would satisfy the data just as well. In geophysical image reconstruction, the solution is generally chosen by imposing some desirable property like smoothness (Sword, 1986), or geological feasibility (van Trier, 1988). The desirable property is generally included by adding a regularisation term to the error function being minimised, creating a new constraint function,

$$
F(d, m)-C(d)+\lambda B(m)
$$

where the vector $\mathbf{d}$ represents the data and $\mathbf{m}$ represents the model parameters. Function C is typically a measure of traveltime residual and function B could be, for example, a smoothing function. In this thesis, C is the sum of squared $X_{\text {err(i) }}$ and a smoothing term (see equation $2.5-1$ ) and B is the entropy function. The parameter $\lambda$ enables trade-off between data fit and model regularisation. This construction results in
a single solution image, but the non-uniqueness of the solution has not really been removed; it is now represented by the arbitrariness of B.

The entropy principle provides a mechanism for choosing, from the range of suitable solution images, the image that has no unjustifiable bias relative to the data and any prior information. Understanding this principle can be confusing as it is possible to derive and explain it from many different viewpoints. Basically, the entropy principle states that the solution image that should be selected is the one with the maximum entropy value (the entropy function will be defined later). The maximum entropy solution can be thought of as the one with a configuration that has the most possible ways of actually occurring. It can also be thought of as the solution which expresses the least amount of information necessary to satisfy the data. Most choices of the regularisation function, $B$, will actually result in solutions that express more information than is required to satisfy the data values and prior information. When this happens, there is an implicit assumption that more information is used than is at hand, and the unwarranted structure of the resulting image cannot be defended. The entropy function guards against this possibility and provides the "common sense" solution.

The history of the entropy principle was developed in two disciplines; thermodynamics and information theory. In thermodynamics, entropy is the basic parameter of the Second Law of Thermodynamics. The history of entropy in thermodynamics has been well documented by Jaynes, 1988a. The entropy function appeared in Kelvin and Clausius' work on heat engines as a measure of the thermodynamic state of the system. Gibbs then showed that any given system, not in equilibrium, will move towards a state of equilibrium that has the maximum entropy allowed by the constraints imposed on the system. Boltzmann explained why
thermodynamic systems approach a state of maximum allowed entropy when he found that the entropy value was actually a measure of the phase volume of a gas with given energy. Since the phase volume corresponds to the number of global quantum states, it is concluded that the maximum entropy state is the one that can occur in the most ways (see Jaynes, 1988a). In thermodynamics, maximum entropy solutions are thought of as states which have the most ways of actually occurring. The concept of least information did not emerge in this context.

As a very simple example, consider the release of some gas molecules into an empty box. What will be the equilibrium distribution of the gas molecules? Common sense suggests that a uniform distribution would result, even though no information exists to allow this conclusion to be deduced (it depends on the initial trajectories and velocities of all the molecules). It appears that there would be many more ways that the gas molecules could physically scatter into a uniform distribution rather than, for example, a distribution where the molecules tend to concentrate in the left half of the box. This is why a uniform distribution is the common sense answer, not because it must be correct but because it has many more chances to be correct. The number of ways any particular distribution can be realised is quantifiable if the box is divided into a number of discrete locations. Given $N$ molecules and $M$ locations, then,

$$
\begin{equation*}
W=\frac{N!}{n_{1}!n_{2}!\ldots n_{M}!} \tag{3.1-1}
\end{equation*}
$$

is the number of ways the molecules can be rearranged to give the same distribution ( $n_{i}$ is the number of molecules at location $i$ ). The value of $W$ is greatest when all $n_{i}$ equal $N / M$, that is, a uniform distribution. If the box is divided into just two halves
and only fifty molecules are considered, then the number of ways a uniform distribution could be realised is,

$$
\frac{50!}{25!25!}-1.26 \times 10^{14},
$$

whereas a distribution with eighty percent of the molecules in one half of the box can only be realised in

$$
\frac{50!}{10!40!}-1.03 \times 10^{10}
$$

ways. Therefore, there are 12233 possible uniform distributions to every $80 \%-20 \%$ distribution. It would be inconsistent (lacking any further information) to choose an $80 \%-20 \%$ distribution as the molecular state when there are so many more ways that the uniform distribution could occur. Furthermore, as the number of molecules increases the uniform distribution becomes even more likely (for 60 molecules the ratio jumps to 84285 to 1 ). When the number of molecules reaches the many millions contained in actual gases, distributions other than uniform become so unlikely that the maximum entropy solution appears to be a law of nature rather than just the most statistically likely solution. It actually turns out that all the classical gas laws can be derived from maximum entropy distributions.

The mathematical link between maximum entropy and the number of ways of realisation is given by the Stirling approximation. If $N$ is large, the Stirling approximation gives (see Jaynes, 1982),

$$
\begin{equation*}
\frac{1}{N} \log W \sim-\sum_{i} \frac{n_{i}}{N} \ln \frac{n_{i}}{N} \tag{3.1-2}
\end{equation*}
$$

which relates the classical entropy function ( $S=-\sum_{i} \frac{n_{i}}{N} \ln \frac{n_{i}}{N}$, see next paragraph) to
the measure of the number of ways that a proposed solution can be realised, $W$. The maximum entropy solution is not necessarily the correct answer, but it has more ways in which it can be correct.

The other discipline to which the principle of entropy can be traced is the field of information theory and the work of Shannon (1947/48 : for an explanation of information theory and proofs of Shannon's theorems, see Khinchin, 1957). The entropy function of information theory has the same form as the right-hand side of equation (3.1-2) and is used as a measure of either information content or the uncertainty of a stochastic distribution. Given a random process with $i$ possible outcomes, each of probability $p_{i}$, then Shannon defined the amount of information gained from the realisation of outcome $i$ to be $-\ln p_{i}$. The mathematical expectation of the information gained is then,

$$
\begin{equation*}
S=-\sum_{i} p_{i} \ln p_{i} \tag{3.1-3}
\end{equation*}
$$

which is once again the classical form of the entropy function.
It would appear that the two historical sources of the entropy function are unrelated, but an information theory analogy can be made to the gas molecule example. Consider a random process that is unknown except for the fact that it has $i$ possible outcomes. What will be the probability distribution of this random process? Common sense suggests that, without any further information, a uniform or flat probability distribution is the best answer. In a very similar way to the gas molecule
example, there are many more ways that the probabilities can be arranged to give a uniform distribution. From the viewpoint of information, any probability distribution which is non-uniform will favour some outcomes over others, but in the absence of prior information the given data does not allow for such conclusions. The uniform distribution represents the least amount of information that needs to be presented in order to solve the problem. Only when further information is available can structure be added to the probability distribution. It is interesting to note that if the mean and variance of a continuous process are known, then the maximum entropy solution is the Gaussian or normal distribution (see Skilling and Gull, 1984).

In a series of important papers, Jaynes (for example 1957, 1982, 1984, 1985, 1988a, 1988b) has explained the rationale and essence of the entropy principle with reference to many disciplines. The entropy principle can be thought of as a means of imposing common sense into difficult problems. As a further simple example, assume that a fair coin has been tossed a million times. What percentage of heads has been tossed? Common sense suggests that fifty percent will be close to correct. This is the maximum entropy answer. However, there is a very slight chance that a million heads had been tossed in a row. Since there are so many more ways that the coin could be tossed to result in fifty percent heads, it makes sense to choose this as the solution (assuming no further information is known). These ideas are related to Laplace's "Principle of Insufficient Reason" (Jaynes (1988b)). In the problems of image reconstruction there are tens of thousands of variables and using the entropy principle ensures that such common sense decisions are made at every step.

During the last few decades, scientists have been applying the principle of entropy to a number of problems outside statistical mechanics and information theory.

Some examples include deconvolution (Skilling and Gull, 1985), spectral estimation (Burg, 1967; Ables, 1974) and various image reconstruction problems in astronomy (again see Skilling and Gull, 1985). Recently the entropy principle has been applied to geophysical traveltime tomography (Bassrei, 1990; Whiting, 1991b).

Image reconstruction using the entropy principle has been used successfully in X-ray, $\gamma$-ray and radio astronomy for a number of years (Gull and Daniell (1978); Skilling and Bryan (1984); Skilling and Gull (1985)). The aim is to reconstruct an image of a portion of the sky from measured radiation. The sky can be divided into discrete cells and the intensity of radiation originating from each cell can be treated as the outcome of a random process (the distribution of stars across the sky is effectively random). Information theory arguments can be used to rationalise the use of entropy in such astronomical problems (even so, there can still be conceptual problems with this, see Cornwell, 1983).

Consider now the problem of traveltime tomographic image reconstruction in geophysics. Here, the objective is to construct an estimate of the subsurface seismic velocity field on a grid of discrete points. It is not obvious how to associate such a problem with any random feature such as the movements of gas molecules or the outcomes of a random process. This introduces a difficulty in the use of entropy constraints in such inverse problems.

One popular way to introduce the entropy principle in image reconstruction problems has been through contingency tables (Gull and Skilling, 1983). The entries of the table are effectively the outcomes of a random process and the application of the entropy principle is straightforward (Good, 1963). Apart from the similar appearance of contingency tables and images divided into regular cells, there is a conceptual
difficulty in making a valid association between the two problems. However, knowledge gained from contingency table analysis is directly applicable to one particularly simple traveltime tomography problem that will be considered in the following sections. Along with a few straightforward rules for image reconstruction, it can be shown that the entropy principle is applicable to general traveltime problems. This results in less spurious structure in the solutions and more stable inversions.

### 3.2 Contingency tables

A two-dimensional contingency table records the number of occurrences of two variables in a crosstabulated format. A typical example is the crosstabulation of the type of weather against the season, as seen in Figure 3.2-1. The entries in the table are the observed number of days of each of the types of weather in each of the four seasons of the year. Analysing a table like this often involves estimating the degree of association between the two variables being observed (see Press et al., 1989, section 13.6), but the primary interest here is in trying to reconstruct the table given only partial information. This problem is ideally suited to the use of the principle of entropy (Good, 1963) since the partial information allows for many possible solutions.

A 'marginal' value of a contingency table is defined as the sum of a single row or column of the table. That is, for example, it represents the total number of fine days or the total number of days in winter. Consider solving for the values of the table given only the eight marginal values. The solution to this problem is highly non-unique and the best solution is not immediately obvious. The marginal distribution for the contingency table is indicated by the arrows of Figure 3.2-1. To demonstrate the nonuniqueness in the problem, the table of Figure 3.2-2 is presented. This reconstructed table is obviously very different to the actual table of Figure 3.2-1. It clearly lacks common sense as well since it suggests that every day in winter was fine and the only days of snow were in spring. Even so, this table does satisfy the marginal distribution. From the marginal values, it rained for 178 days and there were 90 days of winter, however, the reconstruction of Figure 3.2-2 assumes that there were no rainy days


Fig. 3.2-1 A typical contingency table recording the type of weather and the season, sampled over a complete year.
The marginal values are posted next to the arrows.

|  | SEASON |  |  |  |  |  |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: |
| WEATHER | Spring |  | Summ | Autum | Wint |  |
|  | Fine | 0 | 45 | 0 | 90 | $\rightarrow 135$ |
|  | Rain | 43 | 45 | 90 | 0 | $\rightarrow 178$ |
|  | Hail | 19 | 0 | 0 | 0 | $\longrightarrow 19$ |
|  | Snow | 28 | 0 | 0 | 0 | $\rightarrow 28$ |
|  |  | $\begin{aligned} & \downarrow \\ & 90 \end{aligned}$ | $\begin{aligned} & \downarrow \\ & 90 \end{aligned}$ | $\begin{aligned} & \ddagger \\ & 90 \end{aligned}$ | $\begin{aligned} & \ddagger \\ & 90 \end{aligned}$ |  |

Fig. 3.2-2 One of the possible reconstructed contingency tables that satisfy the marginal values.
actually in winter. Given just the marginal data there is no basis for making this claim.
More information is being assumed than is at hand.
The marginal data contains no information about the correlations that exist between any particular row or column. Without any further information, the best that can be done is to assume that the rows and columns have no association and are statistically independent. If the row marginal shows that $p \%$ of the days are rainy, then all that can be said without introducing unjustifiable bias, is that $p \%$ of the days of each season are rainy. If the data are defined as,

$$
r_{i}=\text { row marginal for row } i, i=1,4
$$

and

$$
c_{j}=\text { column marginal for column } j, j=1,4
$$

and with $V_{i j}=$ the reconstructed value crossed by row $i$
and column $j$,
then the unbiased or uncorrelated reconstruction is given by,

$$
\begin{equation*}
V_{i j}=\frac{r_{i} \cdot c_{j}}{N}, \tag{3.2-1}
\end{equation*}
$$

where $N$ is the total number of days (see Press et al., 1989, p478). Reconstructing the table in this way leads to the result of Figure 3.2-3. This result is also very different from the actual table shown in Figure 3.2-1, but it represents all the information we can extract from the marginal data alone without introducing further assumptions. This solution shall be called the "common sense" solution.

It is possible (and usually desirable) to introduce a deliberate bias through the use of prior information. The contingency table in question here is an example where prior information certainly exists; for example, it is known that it is more likely to snow in winter than in summer. To avoid complicating the current analysis the effects of prior information will be considered later.

The usefulness of this contingency table analysis becomes clear when it is observed that the uncorrelated reconstruction (Figure 3.2-3) is actually the maximum entropy reconstruction when given only the marginal data. Frieden (1972) was one of the first to relate entropy to image problems, and Skilling (1988) derived the entropy function particularly for image

| $\begin{gathered} \mathbf{W} \\ \mathbf{E} \end{gathered}$ | Fine | SEASON |  |  |  |  |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: |
|  |  | Spring | Summ | Autum | Wint |  |
|  |  | 33.75 | 33.75 | 33.75 | 33.75 | $\rightarrow 135$ |
|  | Rain | 44.5 | 44.5 | 44.5 | 44.5 | $\longrightarrow 178$ |
| H | Hail | 4.75 | 4.75 | 4.75 | 4.75 | $\longrightarrow 19$ |
|  | Snow | 7 | 7 | 7 | 7 | $\longrightarrow 28$ |
|  |  | $\begin{aligned} & \downarrow \\ & 90 \end{aligned}$ | $\begin{aligned} & \downarrow \\ & 90 \end{aligned}$ | $90$ | $\begin{aligned} & \downarrow \\ & 90 \end{aligned}$ |  |

Fig. 3.2-3 The uncorrelated image that satisfies the marginal values. Created using equation (3.2-1). This is the "common sense" image.
reconstruction. The entropy function for the current contingency table problem can be written as,

$$
\begin{equation*}
S-\sum_{i} \sum_{j}\left(V_{i j}-V_{i j} \ln V_{i j}\right) \tag{3.2-2}
\end{equation*}
$$

The entropy function of equation (3.1-3) is only for values between 0 and 1 (probabilities), this new form does not require normalisation of the image values and is to be maximised directly while satisfying the constraints in the form of the marginal values. The usual method for doing this is by introducing Lagrange multipliers ( $\alpha_{i}$ and
$\beta_{j}$ ) and the variational equation,
$\frac{\partial}{\partial V_{i j}}\left[\sum_{i} \sum_{j}\left(V_{i j}-V_{i j} \ln V_{i j}\right)+\sum_{i} \alpha_{i}\left(\sum_{k} V_{i k}\right)+\sum_{j} \beta_{j}\left(\sum_{k} V_{k j}\right)\right]=0$.
Solving this for one of the locations of the table we get,

$$
\begin{equation*}
-\ln V_{i j}+\alpha_{i}+\beta_{j}=0 \tag{3.2-4}
\end{equation*}
$$

or,

$$
\begin{equation*}
V_{i j}=e^{\alpha_{i}+\beta_{j}}=e^{\alpha_{i}} \cdot e^{\beta_{j}} \tag{3.2-5}
\end{equation*}
$$

where $\alpha_{i}$ and $\beta_{j}$ must be chosen such that the relevant marginal values are satisfied. It is a simple matter to show that equation (3.2-5) is equivalent to equation (3.2-1). Thus, maximising the entropy function (equation (3.2-2)) has reconstructed the unbiased, "common sense" contingency table. The critical feature of the entropy function is the logarithm. The logarithmic form of $S$ directly results in the exponentials in equation (3.2-5), thus converting the added constraints into a multiplicative term that actually allows the multiplication of the marginal values, as required by equation (3.2-1). No other form would be able to convert an addition to a multiplication as required.

Statistical theory allows us to measure the correlation in the table by the use of the covariance,

$$
\begin{equation*}
\operatorname{cov}(i, j)=\mathrm{E}[i j]-\mathrm{E}[i] \mathrm{E}[j] . \tag{3.2-6}
\end{equation*}
$$

Here, the contingency table is viewed as a two-dimensional random variable. The expectation values are defined as,

$$
\left.\begin{array}{c}
\mathrm{E}[i]=\sum_{i} i \frac{R_{i}}{N},  \tag{3.2-7}\\
\mathrm{E}[j]-\sum_{j} j \frac{C_{j}}{N}, \\
\mathrm{E}[i j]-\sum_{i} \sum_{j} i j \frac{V_{i j}}{N} .
\end{array}\right\}
$$

Note that the row and column numbers have been used as the values of the constructed random variables. If the reconstructed image is to be uncorrelated then,

$$
\begin{align*}
\mathrm{E}[i j] & =\mathrm{E}[i] \cdot \mathrm{E}[j] \\
\sum_{i} \sum_{j} i j \frac{V_{i j}}{N} & -\sum_{i} i \frac{R_{i}}{N} \cdot \sum_{j} j \frac{C_{j}}{N}  \tag{3.2-8}\\
& -\sum_{i} \sum_{j} i j \frac{\left(\frac{R_{i} \cdot C_{j}}{N}\right)}{N}
\end{align*}
$$

therefore, $V_{i j}-\frac{R_{i} C_{j}}{N}$, as before. The table created by common sense (Figure 3.2-3)
and by maximising the entropy function is the only uncorrelated reconstruction given the marginal values. Thus the "common sense", uncorrelated and maximum entropy tables are all equivalent.

Maximising entropy (equation (3.2-2)) ensures that no extra information is assumed while the contingency table is being reconstructed. It is desirable to duplicate
this property in general tomographic reconstructions. The next section looks at a simple traveltime tomography problem that is very similar to the contingency table problem. The simple model proves to be very useful later, when more general tomography problems are considered.
3.3 A simple tomographic problem.

Skilling (1988) made use of a simple tomographic problem to demonstrate the necessity of the logarithmic structure of the entropy function. Here, this simple problem provides the link between the above analysis of contingency tables and more general tomographic image reconstruction problems.

The model for this simple problem is a square with unit side length. This square can be evenly divided into any number, $M$ by $M$, of discrete square cells. The aim is to reconstruct the image $V_{i j}$ given measurements of the kind,

$$
\begin{equation*}
\frac{1}{M} \sum_{j=1}^{M} V_{i j}-A_{i}, \quad i-1, \ldots, M \tag{3.3-1}
\end{equation*}
$$

where $1 / M$ is the side length of each cell, and,

$$
\begin{equation*}
\frac{1}{M} \sum_{i=1}^{M} V_{i j}=B_{j}, \quad j=1, \ldots M \tag{3.3-2}
\end{equation*}
$$

That is, there are tomographic projections along every row and down every column. In analogy with the contingency table problem, these data are called marginal data values. Furthermore, assume that the marginal values are normalised,

$$
\begin{equation*}
\sum_{i=1}^{M} A_{i}=\sum_{j=1}^{M} B_{j}=1 \tag{3.3-3}
\end{equation*}
$$

Despite its simplicity, this is a very important problem since the desired reconstructed image is known. From the analysis of contingency tables, the only image that does not introduce unjustified bias is that which multiplies the marginal data values that intersect at each cell. This image is uncorrelated and is the only one which does not assume any
extra information.
The image can now be reconstructed by satisfying the constraints (equations (3.3-1) and (3.3-2)) subject to the maximisation of the entropy function (equation (3.22)). This is done by introducing Lagrange multipliers ( $\alpha_{i}$ and $\beta_{j}$ ) and constructing a variational equation,

$$
\begin{equation*}
\frac{\partial}{\partial V_{i j}}\left[\sum_{i} \sum_{j}\left(V_{i j}-V_{i j} \ln V_{i j}\right)+\sum_{i} \alpha_{i} \sum_{j} \frac{V_{i j}}{M}+\sum_{j} \beta_{j} \sum_{i} \frac{V_{i j}}{M}\right]=0 \tag{3.3-4}
\end{equation*}
$$

which is essentially the same as the variational equation for the contingency table problem (equation (3.2-3)). The reconstructed image for this problem is therefore,

$$
\begin{equation*}
V_{i j}=A_{i} B_{j}, \tag{3.3-5}
\end{equation*}
$$

the product of the marginal data values. Just as with the contingency tables, maximising entropy results in the desired uncorrelated image.

Therefore, given tomographic data of complete simple horizontal and vertical projections through a square medium, it has been shown that maximising entropy during image reconstruction will result in the unique uncorrelated image (as required). Using any function other than the entropy function (equation (3.2-2)) for regularisation may introduce unjustified structure into the image.

The model used in this section serves as the link between contingency tables and tomographic problems. However, tomographic projections in geophysics are curved and not evenly distributed and there are usually far too many of them to permit each to be treated with its own Lagrange multiplier. The next section considers these problems.

### 3.4 General Traveltime Tomography Problems

The general traveltime tomography problem is represented schematically in Figures 3.4-1 and 3.4-2 Figure 3.4-1 displays some typical raypaths that may be encountered in crosswell problems and Figure 3.4-2 shows typical raypaths for reflection data. Note that reflector depths are to be ignored here (see section 2.2). It has been shown that reflection tomography can be performed without reflector depth parameterisation (Sword, 1986; Whiting, 1991b), but combined velocity and depth entropy functions can be constructed if necessary (see Skilling and Gull, 1984). In both the crosswell and reflection cases, the raypaths (or projections) do not form the simple, regular marginal data values of the example in the previous section and the desired solution cannot be obtained by simply multiplying certain data values. The covariance cannot even be measured in these problems as the data are not suitable.


Also, as mentioned earlier, there are generally so many raypaths that it is impractical to assign them individual Lagrange multipliers. ${ }^{2}$ Normally all of the rays are collected together with their errors summed into a single constraint statistic,

$$
\begin{equation*}
C-\sum_{k}\left(t_{d}-t_{m}\right)^{2} \tag{3.4-1}
\end{equation*}
$$

where $t_{d}$ is the actual data traveltime and $t_{m}$ is the traveltime computed by tracing the $k$ th raypath through the model ${ }^{1}$. Some information is actually lost by forming this single $C$ statistic since this statistic does not comprehend where the raypaths are. However, with a good coverage of rays and the assumption that the derivatives, $\frac{\partial C}{\partial V_{i j}}$, can be accurately computed, a useful tomographic reconstruction is still possible. Now that the tomographic problem is specifically defined as a traveltime problem, the image elements, $V_{i j}$, will be referred to as velocity elements (or cell velocities).

The information at hand for the solution of this general tomographic image reconstruction problem is simply the value of the constraint statistic, $C$, and its Frechet derivatives, $\frac{\partial C}{\partial V_{i j}}$. The aim of the tomographic inversion is to minimise $C$, or at least reduce it to a predefined level. Since the function $C$ is highly non-linear, this can only be achieved in practice by making a series of small changes through successive iterations. However, once the predefined level is reached, it is found that a

[^15]large number of images can have the same $C$ value. It is not easy to single out one of these images as any image that reduces $C$ sufficiently is clearly acceptable, given only the constraint statistic and its derivatives. To help choose an image, prior information is introduced. The prior information is supplied from prior knowledge of the medium and is represented by a prior information model, $m_{i j}$, which is essentially the "best guess" starting model. In practice there is always some prior information available, even if it is just a simple velocity increase with depth.

Once the prior model is defined, all that has to be done is to choose the solution which maximises entropy relative to this prior model. With a prior model, the entropy function is modified to (see Skilling, 1988),

$$
\begin{equation*}
S=\sum_{i} \sum_{j}\left(V_{i j}-V_{i j} \ln \frac{V_{i j}}{m_{i j}}\right) . \tag{3.4-2}
\end{equation*}
$$

The form of entropy given in equation (3.2-2) assumed that there was no prior information. Note that equation (3.4-2) reduces to equation (3.2-2) when all $m_{i j}=1$.

However, it is not obvious that maximum entropy solutions will be especially desirable in general tomographic problems. Before a serious analysis of this can proceed, the desirable properties of the solution must be defined. The next section establishes some common sense rules for reconstructing satisfactory solution images.

The general tomographic image reconstruction problem requires a constraint statistic, C, along with its derivatives, $\frac{\partial C}{\partial V_{i j}}$, and a prior information model, $m_{i j}$. The aim of the inversion is to achieve a predefined reduction in $C$ by varying the velocities, $V_{i j}$, from the prior model, $m_{i j}$. The simple example shown in Figure 3.5-1 will be used to help visualise some common sense restrictions to be imposed on the reconstruction of the image. In Figure 3.5-1 the starting position portrayed assumes that $V_{i j}=m_{i j}$ in every cell, thus resulting in $V_{i j} / m_{i j}=1$ everywhere. Also, let the derivative of the constraint statistic be the same for every cell.

There are a large number of ways in which the velocity values can be changed so that the required reduction of the constraint statistic is achieved. At one extreme, a single cell can be changed enough to effect the total necessary reduction (see Figure 3.5-2). However the choice of the modified cell is totally arbitrary. Such sources of arbitrariness are contrary to the common sense approach of maximum entropy methods.


Fig. 3.5-1 The starting position of an instructive image reconstruction problem. $\mathrm{V}_{\mathrm{ij}} / \mathrm{m}_{\mathrm{ij}}$ is plotted where, at the start, $\mathrm{V}_{\mathrm{ij}}=\mathrm{m}_{\mathrm{ij}}$ everywhere.

Since all of the constraint statistic derivatives are equal, common sense suggests that no cell can be singled out in preference to any other cell. In this case, the only sensible reconstruction is the one which changes all of the cell velocities equally. Picking a single cell to carry the full burden of the $C$ reduction causes the reconstruction to make a very strong commitment to that cell. However, it is


Fig. 3.5-2 A single cell is changed sufficiently (to $\mathrm{V}_{\mathrm{ij}} / \mathrm{m}_{\mathrm{ij}}=\delta$ ) to reduce C to the required level. preferable to impose as little commitment to any one cell as possible while reducing $C$ by the specified amount. Intuitively, the reconstruction with equal changes that is desired here, would also be the reconstruction with the least overall commitment.

On the other hand, if one cell had a larger constraint statistic derivative than the rest, it would be desirable for this cell to bear more of the required velocity change. At first it appears that this would increase the overall commitment level, but because of the larger derivative, the velocity change in this cell would be more than offset by a decrease of the change required in all the other cells. The final result would actually be of lower overall commitment than for a uniformly changed model. Clearly, for the concept of commitment to be evaluated further, a method of measuring the overall level of commitment must be defined.

Assume that it is possible to measure the amount of commitment using a nonnegative function, $\Psi$, of $V_{i j} / m_{i j}$ (let $x_{i j}=V_{i j} / m_{i j}$ for brevity). It is proposed that the
following four properties suffice to define such a function (with $m_{i j}>0$, and $\psi$ defined for $x_{i j}>0$ );
(a) $\psi\left(x_{i j}\right) \geq 0$,
(b) $\psi(1)=\psi^{\prime}(1)-0$,
(c) $\psi^{\prime \prime}\left(x_{i j}\right)>0 \quad$ for all $x_{i j}$,
and (d) $\psi^{\prime}\left(x_{i j}\right) \rightarrow-\infty$ as $x_{i j} \rightarrow 0^{+}$.

Property (a) implies that any variation of $V_{i j}$ from $m_{i j}$ introduces positive commitment.

Property (b) states that $V_{i j}=m_{i j}$ is the state of zero commitment and that this is a local minimum. Properties (c) and (d) will require further explanation.

To help understand property (c), imagine that Figure 3.5-2 is an intermediate reconstruction (even though not a very acceptable one) and all of the gradients, $\frac{\partial C}{\partial V_{i j}}$, are equal. To reduce $C$ to the required level assume that a further increment $\in(>0)$ in velocity is required. Figures $3.5-3$ and $3.5-4$ suggest two of the possible ways of doing this, either by adding the extra change to the cell already changed or adding it to any other cell. The image of Figure $3.5-3$ has greater overall commitment than that of Figure 3.5-4, and for the function $\psi$ to reflect this,

$$
[\Psi(\delta+\varepsilon)-\Psi(\delta)]>[\Psi(1+\varepsilon)-\Psi(1)]
$$

or more generally,

$$
[\Psi(a+\varepsilon)-\psi(a)]>[\psi(b+\varepsilon)-\psi(b)] \quad, \quad \text { if } a>b \geq 1 .
$$

| 1 | 1 | 1 | 1 | 1 |
| :---: | :---: | :---: | :---: | :---: |
| 1 | 1 | 1 | $\delta+\epsilon$ | 1 |
| 1 | 1 | 1 | 1 | 1 |
| 1 | 1 | 1 | 1 | 1 |


|  |  |  |  |  |
| :---: | :---: | :---: | :---: | :---: |
| 1 | 1 | 1 | 1 | $1+\epsilon$ |
| 1 | 1 | 1 | $\delta$ | 1 |
| 1 | 1 | 1 | 1 | 1 |
| 1 | 1 | 1 | 1 | 1 |

Fig. 3.5-4 The extra velocity change $\epsilon$ could be added to any other cell.

Fig. 3.5-3 The extra velocity change $\epsilon$ could be added to the cell already changed.

This inequality is fulfilled if $\psi$ is convex, or $\psi^{\prime \prime}>0$ for all $x_{i j}$.

The above inequality is actually more powerful than it first appears. If $\varepsilon$ is to be added to one of the cells in Figure 3.5-4 other than the cell of value $\delta$, then the above inequality gives,

$$
\begin{gathered}
{\left[\Psi\left(\left[1+\frac{\epsilon}{2}\right]+\frac{\epsilon}{2}\right)-\Psi\left(1+\frac{\epsilon}{2}\right)\right]>\left[\Psi\left(1+\frac{\epsilon}{2}\right)-\Psi(1)\right]} \\
o r, \\
\Psi(1+\epsilon)>2 \Psi\left(1+\frac{\varepsilon}{2}\right)
\end{gathered}
$$

This means that splitting the additional $\varepsilon$ increment into two halves and adding these to two cells, is a state of less commitment than adding all of $\varepsilon$ to a single cell. This argument can obviously be carried on until the $\varepsilon$ velocity increment is evenly divided across all available cells. It can be shown that when the original $\delta$ increment was
made (Figure 3.5-2), the minimum commitment reconstruction would have been one with equal velocity changes in every cell. Therefore, the simple condition of convexity imposed on the commitment function, $\psi$, is sufficient to ensure that least commitment reconstructions will be sensible, as intuitively defined earlier.

The convexity property defined above, also has relevance for the case where the constraint statistic derivatives are not equal. In this case, common sense requires that more of the velocity change be carried by cells with larger derivatives. For the moment, consider a two velocity cell example where the constraint statistic derivatives are not equal (also assume that both derivatives are negative - similar arguments follow for other values). The reconstruction with equal changes in velocity must not be the least commitment reconstruction in this case. Due to the convexity of the commitment function, an increase in the velocity of one cell will only produce lower commitment if the decrease in the other cell is greater (how much greater depends on the degree of convexity). The only way this can happen, and still produce the same constraint statistic reduction, is if the cell with the larger derivative is the one increased. Therefore, the convexity of the commitment function also forces more of the velocity change to be taken up in cells with larger constraint statistic derivatives.

The fourth property, where the derivative of the commitment function is asymptotic to $-\infty$ at the $\mathrm{x}=0$ axis, is specified in order to make it effectively impossible for a zero velocity to be reconstructed. Negative and zero velocities will be naturally avoided.

The four properties listed above define a commitment function that will enforce apparent common sense (as intuitively described above) into the reconstruction of the image. However, the definition is not so constrained as to disallow quite a degree of
flexibility. The form of the required regularisation function has been defined, but not its details. Skilling (1988) showed that if the reconstruction problem happened to be of the form described in the simple tomographic problem section, then the regularisation function needed to obtain the required solution is the entropy function of equation (3.42). This special case implies a commitment function given by,

$$
\begin{equation*}
\Psi\left(x_{i j}\right)-A\left[1-x_{i j}+x_{i j} \ln x_{i j}\right] \quad ; A>0, \tag{3.5-1}
\end{equation*}
$$

which satisfies all four properties described previously. The fact that the entropy function is maximised in practice rather than $\psi$ minimised, means that entropy can be thought of as a measure of non-commitment.

The entropy function (equation (3.4-2)) is schematically represented in Figure 3.5-5. It is maximum at $x_{i j}=V_{i j} / m_{i j}=1$ and its domain is $x_{i j} \geq 0$. Also, the logarithmic form ensures that the magnitude of its derivative monotonically increases as $V_{i j} / m_{i j}$ moves away from 1 (hence satisfying property (c)). Therefore, based on the preceding discussion, maximising the entropy function will ensure that commitment to individual cells is kept to a minimum. This in turn implies that the entropy function will ensure sensible (or common sense) image reconstructions in both the "simple" and "general" tomography problems.

The actual form of the maximum entropy reconstruction is most easily observed from the variational equation. The general problem is to minimise the constraint statistic $C$ while maximising entropy. In practice however it is more convenient to maximise entropy subject to an acceptable level of $C$. This reduces to the variational equation,


Fig. 3.5-5 The entropy function (equation (3.4-2)) represented in graphical form for a single cell.

$$
\begin{equation*}
\frac{\partial}{\partial V_{i j}}\left[\sum_{i} \sum_{j}\left(V_{i j}-V_{i j} \ln \frac{V_{i j}}{m_{i j}}\right)+\lambda C\right]-0, \tag{3.5-2}
\end{equation*}
$$

where $\lambda$ is a (single) Lagrange multiplier. This leads to,

$$
-\ln \frac{V_{i j}}{m_{i j}}+\lambda \frac{\partial c}{\partial V_{i j}}=0
$$

or, equivalently,

$$
\begin{equation*}
\frac{V_{i j}}{m_{i j}}=e^{\lambda \frac{\partial c}{\partial V_{j}}}, \tag{3.5-3}
\end{equation*}
$$

where $\lambda$ is chosen so that the required reduction of C is achieved. Equation (3.5-3) implies that all the cells with the same gradient, $\partial C / \partial V_{i j}$, will have exactly the same velocity change relative to their prior values, $m_{i j}$. Also, the larger the value of $\partial C / \partial V_{i j}$, the larger will be the relative velocity change. Some simple examples will now show how maximising entropy influences actual inversions.

### 3.6 Simple Examples

In this section, the common sense properties (defined in the previous section) of desirable tomographic image reconstructions will be demonstrated with the help of some very simple examples. Maximum entropy reconstructions will be compared with three other possible solution images. The first reconstructed image that will be compared to the maximum entropy image is just the smoothest image that solves the constraints. The second is the "least change" solution ${ }^{1}$ (that is, the one that minimises $\sum\left(v_{i j}-v_{i j}^{o}\right)^{2}$ where $v_{i j}^{o}$ is the starting velocity), and the third is the "least change from prior" solution (the one that minimises $\sum\left(v_{i j}-m_{i j}\right)^{2}$ where $m_{i j}$ is the prior information velocity model).

The first example demonstrates the case where the reconstructed image may not be representative of the information given by the constraint statistic derivatives, or more simply, the sensitivities. Consider an image that is to be tomographically reconstructed and consists of only two cells. This model could be traversed by any number of rays but the only information that matters is the sensitivity of the constraint statistic to velocity changes in each cell. For this example, assume that the sensitivities are $\frac{\partial C}{\partial v_{1}}=-1$ and $\frac{\partial C}{\partial v_{2}}=-2$, and that the required $C$ reduction for an acceptable solution is 1 unit. With a starting model equal to the prior information model of $v_{1}=v_{2}=1$, the

[^16]$v_{1}=v_{2}=1$, the reconstructions are $\left(v_{1}, v_{2}\right)=(1.1861,1.4069)$, ( $1.3333,1.3333$ ), $(1.2,1.4)$ and $(1.2,1.4)$, being respectively the maximum entropy, smoothest, least change, and least change from prior solutions. If the sensitivities are reliable then the smoothest solution does not honour their relative magnitudes. For this example, the least change images and the maximum entropy image are quite similar and, considering the sensitivities, appear to have made sensible reconstructions. If prior knowledge is available that suggests that the image should be smooth to some degree, this can be built into the constraint statistic and the maximum entropy principle can still be applied.

The next example is slightly more complex and demonstrates two important features of sensible reconstructions. Consider the nine cell, 2 -dimensional problem of Figure 3.6-1. For the required $C$ reduction of 9 units, the 4 chosen solutions have been generated (see Figure 3.6-2). The maximum entropy image (Figure 3.6-2(d)) is perfectly smooth, as it should be since each cell has exactly the same sensitivity and the prior information model is smooth. The least change solution (Figure 3.6-2(b)) maintains
 method is too greatly affected by earlier changes of the velocities. This is a serious shortfall since it is common for cells to be erroneously changed by early iterations. The top right cell may have been updated by an earlier iteration which had its rays incorrectly placed, and the least-squares minimum change criterion (which is the criterion used by the popular SIRT algorithm (see Stewart, 1989)) is incapable of correcting this as the ray locations improve. Such incorrect updates will appear as spurious additional structure on the final image. For this example, both the least change from prior and smoothest solutions (Figures 3.6-2(c) and 3.6-2(a)) are identical


Fig. 3.6-1 A nine element 2D problem. (a) is the starting (or intermediate) model, (b) is the prior information model, and (c) contains the sensitivities of the constraint statistic. The required reduction of C is 9 units.

| 2.22 | 2.22 | 2.22 |
| :--- | :--- | :--- |
| 2.22 | 2.22 | 2.22 |
| 2.22 | 2.22 | 2.22 |

(a)

| 2.22 | 2.22 | 2.22 |
| :--- | :--- | :--- |
| 2.22 | 2.22 | 2.22 |
| 2.22 | 2.22 | 2.22 |

(c)

| 2 | 2 | 4 |
| :--- | :--- | :--- |
| 2 | 2 | 2 |
| 2 | 2 | 2 |

(b)

| 2.22 | 2.22 | 2.22 |
| :--- | :--- | :--- |
| 2.22 | 2.22 | 2.22 |
| 2.22 | 2.22 | 2.22 |

(d)

Figure 3.6-2 Four possible solutions to the problem of Figure 3.6-1. (a) is the smoothest solution, (b) is the least squares minimum change solution, (c) has the minimum change from the prior, and (d) is the maximum entropy image.

A further two-dimensional example is depicted in Figure 3.6-3. On this occasion the sign of the sensitivities create negative velocity values for all reconstructed images except the maximum entropy image (see Figure 3.6-4). The negative values can be avoided by imposing a minimum velocity (eg. Carrion et al., 1993), but this would be just another source of arbitrariness. The maximum entropy solution guarantees positivity.

In all of the above examples the prior information model was a uniform one and so did not supply any structural information as the prior velocity of each cell was set to unity. To see how prior information can help with the reconstruction, consider the problem of Figure 3.6-5. The starting model here is set equal to the prior model and suggests a velocity increase with depth (see Figure 3.6-5(a)). Given the sensitivities of Figure 3.6-5(b), the maximum entropy image is shown in Figure 3.6-6. Even though the cell at the bottom of the model had a larger positive sensitivity (positive sensitivities will reduce $V_{i j}$ ), its final velocity is larger than that of the cell at the top of the model because of the different prior values. One way to view the use of prior information is to notice that the maximum entropy reconstruction is not concerned with the actual velocity values, but rather with the ratio of velocity and prior model velocity, $V_{i j} / m_{i j}$ (see equation (3.5-3)). Alternatively, rearranging equation (3.5-3),

$$
\begin{align*}
V_{i j} & =m_{i j}\left(\mathrm{e}^{\left.\frac{\partial C}{}\right)^{\frac{\partial V_{i}}{\partial y}}}\right.  \tag{3.6-1}\\
& -\operatorname{prior}(\text { const })^{\text {sensitivity }}
\end{align*}
$$

Once $\lambda$ is chosen to satisfy the constraints, $e^{\lambda}$ becomes a constant for the whole image. The magnitude and sign of the cell's sensitivity then exponentiates this

| 1 | 1 | 1 |
| :--- | :--- | :--- |
| 1 | 2 | 2 |
| 1 | 1 | 1 |

(a)

| 1 | 1 | 1 |
| :--- | :--- | :--- |
| 1 | 1 | 1 |
| 1 | 1 | 1 |

(b)

| 2 | 0 | 0 |
| :---: | :---: | :---: |
| 0 | -1 | 0 |
| 0 | 0 | 0 |

(c)

Fig. 3.6-3 Another nine element 2D problem. (a) is the starting (or intermediate) model, (b) is the prior information model, and (c) contains the sensitivities of the constraint statistic, C . The required C reduction is 8 units.

| -8 | -8 | -8 |
| :--- | :--- | :--- |
| -8 | -8 | -8 |
| -8 | -8 | -8 |

(a)

| -2.6 | 1 | 1 |
| :---: | :---: | :---: |
| 1 | 2.8 | 1 |
| 1 | 1 | 1 |

(c)

| -2.2 | 1 | 1 |
| :---: | :---: | :---: |
| 1 | 3.6 | 2 |
| 1 | 1 | 1 |

(b)

| .016 | 1 | 1 |
| :---: | :---: | :---: |
| 1 | 8.03 | 1 |
| 1 | 1 | 1 |

(d)

Fig. 3.6-4 Four possible solutions to the problem outlined in Figure 3.6-3. (a) the smoothest, (b) least-squares minimum change, (c) nearest to prior, and (d) maximum entropy.
constant, and the prior value is a simple multiplier. As a cell sensitivity approaches zero (little or no information) then $V_{i j}$ approaches the prior value. The prior acts as both a default value and as a relative scaler for the image values. Relevant prior information model values can help keep the inversion in good shape through the iterations, especially when some poor sensitivity estimates appear.


Fig. 3.6-5 A problem with a prior information model that suggests a velocity increase with depth. (a) is the starting and prior model, (b) contains the sensitivities of the constraint statistic. The required C reduction is 4 units.


Fig. 3.6-6 The maximum entropy image for the problem of Figure 3.6-5

### 3.7 Synthetic Data Results

Figure $3.7-1$ is the velocity/depth model used in a synthetic crosswell tomography problem. Sources were located at a 20 metre spacing down the left side of the model and receivers were similarly spaced down the right side, resulting in 676 raypaths in total. The data were initially inverted with the conventional SIRT algorithm (see Stewart, 1989) which finds the solution image with least-squares minimum change. The SIRT reconstruction is shown in Figure 3.7-2. The corresponding maximum entropy reconstruction, using a constant $2000 \mathrm{~m} / \mathrm{s}$ prior model, is shown in Figure 3.7-3. Notice how all 3 anomalies may be more apparent in the maximum entropy reconstruction (the maximum entropy inversion algorithm used is based on that of Skilling and Bryan, 1984; see section 4). Also, the velocities are nearer to the correct values. Away from the anomalies, the maximum entropy image also exhibits less


Fig. 3.7-1 Velocity/depth model for a synthetic crosswell tomography problem. Sources and receivers separated by 20 metres; total of 676 raypaths.


Fig. 3.7-2 The SIRT reconstructed image for the crosswell synthetic model of Figure 3.7-1


Fig. 3.7-3 The maximum entropy reconstructed image for the crosswell synthetic model of Figure 3.7-1
spurious structure since each maximum entropy iteration effectively questions the changes made in earlier iterations. If there is no longer justification for the change, it tends to reset the velocity of the cell back to that of the prior model.

An example of reflection tomography (Whiting, 1991b), both with and without maximum entropy constraints, will be demonstrated using the model of Figure 3.7-4. This model consists of three constant velocity layers with a high angle fault roughly in the centre of the model. The starting model and the prior information model consisted of the three layers without the fault and with a smooth velocity transition from one layer to the next. The results of two inversions can be seen in Figures 3.7-5 and 3.7-6, the former being the reconstruction without any entropy consideration and the latter being the maximum entropy image. Notice that the maximum entropy image has less unwarranted structure. The only "bumps" occurring in the layer boundaries are those associated with the faults.


Fig. 3.7-4 Velocity/depth model for the synthetic reflection tomography example. 800 raypaths were used with random source/receiver locations and random reflection points.


Fig. 3.7-5 The image reconstructed for the synthetic reflection tomography example without entropy constraints.


Fig. 3.7-6 The maximum entropy image reconstructed for the synthetic reflection tomography example.

### 3.8 Discussion

The entropy principle has a long and successful history. The same theory as developed in statistical mechanics and information theory can be used, with suitable adjustments, in tomographic image reconstruction. The reason for the universality is that the entropy principle is basically an expression of "common sense". When faced with a range of possible solutions, entropy considerations suggest that the solution that has the most ways of actually occurring should be chosen. If a solution is chosen that has fewer ways of actually occurring, then it is implicit that more information is being assumed. This notion is consistent with the concept of "least commitment" in tomographic image reconstruction. Choosing an image with more commitment to individual cells, or structure, implicitly assumes that more information is known. Maximising entropy ensures that only as much information as is possible (given the current prior information) is extracted from the data, and no more.

The entropy function (introduced in tomographic image reconstruction with the assistance of contingency table analysis) ensures that the velocities are always positive and that the reconstructed values reflect both the size of the constraint statistic sensitivities and the structure of the prior information model. In this way, as long as the sensitivities are mostly noise free and the inversion is not trapped by a local minimum, a useful and relevant image can be reconstructed. Examples of reconstructions of a crosswell and a reflection traveltime tomographic inversion have demonstrated these features and the usefulness of the entropy principle in tomographic image reconstruction.

## §4 THE ENTROPY PRINCIPLE - APPLICATION

4.1 The basic maximum entropy inversion algorithm.

Inversion of traveltime data by the entropy principle attempts to minimise a residual traveltime error statistic while maintaining maximum entropy. This is a simultaneous optimisation problem. Such a problem is of considerable complexity because the model space is an $n$-dimensional space commonly of much more than 10,000 dimensions. Computational efficiency is paramount under these circumstances and matrix techniques are not feasible. The algorithm used in this study is based on that described by Skilling and Bryan (1984). They developed a general algorithm but their primary objective was astronomical imaging. This section outlines the basic strategy of this algorithm. The following sections will describe the additions and modifications that have been made for the current study.

The general problem to be solved by the maximum entropy inversion algorithm is to find model parameters, $\mathbf{q}$, that best predict the given data, d , given a theoretical relationship (generally non-linear) of the form,

$$
\begin{equation*}
d=\mathbb{R}(q) \tag{4.1-1}
\end{equation*}
$$

Here, d represents an $m$-vector of data values, q represents an $n$-vector of model parameters, and $\mathbb{R}$ represents a mapping from model space to the data space. Generally, this non-linear operator $\mathbb{R}$ is linearised and the inversion process is
performed in an iterative fashion. In this case, equation (4.1-1) can be written as,

$$
\begin{equation*}
\delta d_{i}-\sum_{j} R_{i j} \delta q_{j} \tag{4.1-2}
\end{equation*}
$$

a standard linear matrix equation, where the matrix $R_{i j}$ represents the linear approximation of $\mathbb{R}$, and the $\delta d_{i}, \delta q_{j}$ notation implies that this linear approximation is generally only valid for small changes of the parameters. As the problem is solved in a series of small steps, a constraint statistic, $C$, is introduced to ensure that the model parameters are continually being changed toward those that most accurately predict the data. The simplest form of this constraint statistic is just the data residual,

$$
\begin{equation*}
C=(\mathrm{d}-\mathbb{R}(\mathrm{q}))^{\mathrm{T}} C_{d}^{-1}(\mathrm{~d}-\mathbb{R}(\mathrm{q})) \tag{4.1-4}
\end{equation*}
$$

(which assumes Gaussian statistics) where $C_{d}$ is the data covariance matrix. The iterative technique then attempts to reduce $C$ at every iteration until the minimum is located. If the number of data and model parameters are small, the matrix $R_{i j}$ will have reasonable dimension and it is feasible to invert it using standard linear algebra. The inverted matrix can then be used to give the updated model parameters for each iteration directly,

$$
\begin{equation*}
R^{-1} \delta \mathrm{~d}-\delta \mathrm{q} \tag{4.1-3}
\end{equation*}
$$

However, the problem is rarely well conditioned and the matrix is often not invertible.
Also, the size of the $R$ matrix quickly makes matrix operations impractical.
The role of the inversion is to find the global minimum of a non-linear function (ie. the constraint statistic, $C$ ) within an $n$-dimensional space, where ' $n$ ' is very large ( $>10,000$ ). An iterative approach to finding this global minimum begins with a set of
initial model parameters which defines a single point in this n-space. At this location, it is necessary to estimate $\nabla C$ and construct a local linear approximation to the $C$-surface that can be used to move to a nearby point with a (hopefully) smaller value of $C$. A new local linear approximation is made and the next small step is taken. The procedure iterates in this fashion until a minimum of the constraint statistic is located. Optimisation algorithms that work in this way are called 'descent methods'. Commonly used variants are the 'steepest descent' and 'conjugate gradient' techniques.

If the second-order gradient of $C, \nabla \nabla C$, can be estimated, the linear approximation can be upgraded to a locally quadratic approximation. Quadratic approximations help provide reasonable limits on the iterative step distance. With just $\nabla C$ it is difficult to decide on the size of model parameter change that can be accommodated by each iteration. It is generally difficult to compute $\nabla \nabla C$ but approximations to it also prove to be useful. The conjugate gradient technique attempts to build up information about $\nabla \nabla C$ as the iterations proceed.

Most descent methods have a criterion for choosing a single search direction in n-space based on $C, \nabla C$ and possibly $\nabla \nabla C$ at the current location. In between each iteration, new data estimates are computed using equation (4.1-1) and $\nabla C$ and $\nabla \nabla C$ must be re-evaluated. This inter-iteration computing is significant and it is more efficient and sensible to use more than just one search direction for each iteration. This suggests searching in a subspace rather than along a single line. Subspace inversion methods are not new in geophysics (see Kennett et al., 1988). The results of a subspace search can be no worse than a linear search as the direction of linear search
is included within the subspace. Skilling and Bryan (1984) suggest a 4 or 6 dimensional subspace.

Optimisation with the maximum entropy principle requires the minimisation of C while maximising the entropy function, $S$ (see section 3). Skilling and Bryan (1984) suggest simultaneous optimisation of these functions ${ }^{1}$. Successful simultaneous optimisation relies on the subspace to provide reasonable approximations of both $S$ and C. Inspired by the conjugate gradient method, Skilling and Bryan (1984) choose the following six vectors for the definition of the subspace,

$$
\begin{gather*}
\nabla S, \nabla \nabla C . \nabla S, \nabla \nabla C . \nabla \nabla C . \nabla S  \tag{4.1-5}\\
\nabla C, \nabla \nabla C . \nabla C, \nabla \nabla C . \nabla \nabla C . \nabla C
\end{gather*}
$$

Even though $\nabla \nabla C$ is potentially a very large matrix, each of these vectors can be derived by vector operations only.

Note that $\nabla \nabla S$ does not appear in these definitions (4.1-5). This is because Skilling and Bryan (1984) introduce an "entropy metric". In their astronomical application the model parameters can have very small values tending toward zero. To help avoid some of these parameters assuming negative values Skilling and Bryan measure distances as,

$$
\begin{equation*}
\sum_{i} \frac{\delta q_{i}^{2}}{q_{i}} \tag{4.1-6}
\end{equation*}
$$

where $\delta q_{i}$ represents a change of the $i$ th model parameter. This is equivalent to a

[^17]metric of $\frac{1}{q_{i}}$ which can be shown to be equal to $-\nabla \nabla S$. Skilling and Bryan call this the entropy metric. This metric is of little importance in subsurface interval velocity imaging since the dynamic range of velocity variations is quite small. The effect of using this metric is to modify the subspace vectors (4.1-5) to,
\[

$$
\begin{gather*}
q(\nabla S), q(\nabla \nabla C . \nabla S), q(\nabla \nabla C . \nabla \nabla C . \nabla S)  \tag{4.1-7}\\
q(\nabla C), q(\nabla \nabla C . \nabla C), q(\nabla \nabla C . \nabla \nabla C . \nabla C)
\end{gather*}
$$
\]

where the $q()$ notation implies component by component multiplication. Now it is clear that $q(\nabla \nabla S)=I$ and would add nothing if included in the subspace vector definition. These six vectors (4.1-7) define the subspace to be used, however, it is common to use only a 4-dimensional subspace with the first two vectors of each type.

The subspace vectors can be organised into an ( $\mathrm{n} \times \mathrm{r}$ )-matrix, E , that maps vectors from the original $n$-dimensional space to the $r$-dimensional subspace,

$$
\begin{equation*}
E^{T} q-\mathbf{y} \tag{4.1-8}
\end{equation*}
$$

In the $n$-dimensional model space, the original image $\mathbf{q}_{0}$ is to be updated to a new image $\mathbf{q}$,

$$
\begin{equation*}
q-q_{0}+\delta q \tag{4.1-9}
\end{equation*}
$$

The constraint statistic, $C$, and the entropy, $S$, surfaces can be expressed as, $S=S_{0}+\boldsymbol{\delta} \mathbf{q}^{\mathbf{T}} \nabla S_{0}+\frac{1}{2} \boldsymbol{\delta} \mathbf{q}^{\mathbf{T}} \nabla \nabla S_{0} \mathbf{\delta} \mathbf{q}+$ higher order terms
$C=C_{0}+\delta \mathbf{q}^{\mathbf{T}} \nabla C_{0}+\frac{1}{2} \delta \mathbf{q}^{\mathbf{T}} \nabla \nabla C_{0} \delta \mathbf{q}+$ higher order terms
where $C_{0}$ and $S_{0}$ are the values of $C$ and $S$ at $\mathbf{q}_{0}$. Equation (4.1-8) provides the
${ }^{5} D_{\beta}=\operatorname{diag}\left[\beta_{1}, \beta_{2}, \ldots\right]$ where $\beta_{1}, \beta_{2}, \ldots$ are the eigenvalues of $B$.
means for mapping these surfaces to the $r$-dimensional subspace. If the higher order terms of equations (4.1-10) and (4.1-11) are ignored, quadratic approximations result,

$$
\begin{align*}
& \bar{S}(\mathrm{y})-S_{0}+y^{\mathrm{T}} \mathrm{a}-\frac{1}{2} \mathrm{y}^{\mathrm{T}} \mathrm{Ay}  \tag{4.1-12}\\
& \bar{C}(\mathrm{y})-C_{0}+y^{\mathrm{T}} \mathrm{~b}+\frac{1}{2} \mathrm{y}^{\mathrm{T}} \mathrm{By} \tag{4.1-13}
\end{align*}
$$

where,

$$
\begin{aligned}
& \mathrm{a}-\mathrm{E}^{\mathrm{T}} \nabla S_{0}, \mathrm{~A}-\mathrm{E}^{\mathrm{T}}\left(-\nabla \nabla S_{0}\right) \mathrm{E}, \\
& \mathrm{~b}=\mathrm{E}^{\mathrm{T}} \nabla C_{0}, \mathrm{~B}=\mathrm{E}^{\mathrm{T}}\left(\nabla \nabla C_{0}\right) \mathrm{E}
\end{aligned}
$$

These approximations are the basis of the simultaneous minimisation of the constraint statistic and maximisation of entropy.

The mathematical manipulation of the quadratic approximations within the subspace (equations (4.1-12) and (4.1-13)) is simplified if the matrices A and B are simultaneously diagonalised. A detailed examination of this process is left for section 4.2. As will be seen in section 4.2, a transformation is applied to the subspace such that equations (4.1-12) and (4.1-13) become,

$$
\begin{align*}
& \bar{S}(\mathrm{x})=S_{0}+\mathrm{x}^{\mathrm{T}} \mathrm{a}^{\prime}-\frac{1}{2} \mathrm{x}^{\mathrm{T}} \mathrm{x}  \tag{4.1-14}\\
& \bar{C}(\mathrm{x})=C_{0}+\mathrm{x}^{\mathrm{T}} \mathrm{~b}^{\prime}+\frac{1}{2} \mathrm{x}^{\mathrm{T}} \mathrm{D}_{\beta} \mathrm{x}, \tag{4.1-15}
\end{align*}
$$

where $\mathrm{x}, \mathrm{a}^{\prime}$ and $\mathrm{b}^{\prime}$ are the transformed versions of y , a and b respectively, and $\mathrm{D}_{\beta}$ is a diagonal matrix. Notice that $\mathbf{A}$ has been transformed to the identity matrix and $\mathbf{B}$ has been diagonalised. These equations simplify the mathematics considerably.

It is now possible to locate the minimum value of the quadratic approximation, $\bar{C}(\mathrm{x})$, with respect to the $r$-dimensional subspace. Set the derivative with respect tox
means for mapping these surfaces to the $r$-dimensional subspace. If the higher order terms of equations (4.1-10) and (4.1-11) are ignored, quadratic approximations result,

$$
\begin{align*}
& \bar{S}(\mathrm{y})=S_{0}+\mathrm{y}^{\mathrm{T}} \mathrm{a}-\frac{1}{2} \mathrm{y}^{\mathrm{T}} \mathrm{Ay}  \tag{4.1-12}\\
& \bar{C}(\mathrm{y})=C_{0}+\mathrm{y}^{\mathrm{T}} \mathrm{~b}+\frac{1}{2} \mathrm{y}^{\mathrm{T}} \mathrm{By} \tag{4.1-13}
\end{align*}
$$

where,

$$
\begin{aligned}
& \mathrm{a}=\mathrm{E}^{\mathrm{T}} \nabla S_{0}, \mathrm{~A}=\mathrm{E}^{\mathrm{T}}\left(-\nabla \nabla S_{0}\right) \mathrm{E}, \\
& \mathrm{~b}-\mathrm{E}^{\mathrm{T}} \nabla C_{0}, \mathrm{~B}=\mathrm{E}^{\mathrm{T}}\left(\nabla \nabla C_{0}\right) \mathrm{E}
\end{aligned}
$$

These approximations are the basis of the simultaneous minimisation of the constraint statistic and maximisation of entropy.

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$$
\begin{align*}
& \bar{S}(\mathrm{x})-S_{0}+\mathrm{x}^{\mathrm{T}} \mathrm{a}^{\prime}-\frac{1}{2} \mathrm{x}^{\mathrm{T}} \mathrm{x}  \tag{4.1-14}\\
& \bar{C}(\mathrm{x})=C_{0}+\mathrm{x}^{\mathrm{T}} \mathrm{~b}^{\prime}+\frac{1}{2} \mathrm{x}^{\mathrm{T}} \mathrm{D}_{\beta} \mathrm{x}, \tag{4.1-15}
\end{align*}
$$

where $x, a^{\prime}$ and $b^{\prime}$ are the transformed versions of $y, a$ and $b$ respectively, and $D_{\beta}$ is a diagonal matrix. Notice that $\mathbf{A}$ has been transformed to the identity matrix and $\mathbf{B}$ has been diagonalised. These equations simplify the mathematics considerably.

It is now possible to locate the minimum value of the quadratic approximation,

[^18]to zero,
$$
\frac{\partial \bar{C}}{\partial \mathbf{x}}-\mathbf{b}^{\prime}+\mathrm{D}_{\beta} \mathrm{x}=0 .
$$

Therefore,

$$
\begin{equation*}
\mathbf{X}_{\min }=\frac{-\mathbf{b}^{\prime}}{D_{\beta}}=-D_{\frac{1}{\beta}} b^{\prime} \tag{4.1-16}
\end{equation*}
$$

is the subspace location of the constraint statistic minimum. This can be substituted into equation (4.1-15) to give the actual value of the constraint statistic minimum,

$$
\begin{align*}
\bar{C}_{\min } & =C_{0}+\left(-\mathbf{b}^{\prime} \mathrm{D}_{\frac{1}{\beta}}\right)^{\mathrm{T}} \mathbf{b}^{\prime}+\frac{1}{2}\left(-\mathbf{b}^{\prime} \mathrm{D}_{\frac{1}{\beta}}\right)^{\mathrm{T}} \mathrm{D}_{\beta}\left(-\mathbf{b}^{\prime} \mathrm{D}_{\frac{1}{\beta}}\right) \\
& =C_{0}-\mathbf{b}^{\prime \mathbf{T}} \mathrm{D}_{\frac{1}{\beta}} \mathbf{b}^{\prime}+\frac{1}{2} \mathbf{b}^{\prime \mathrm{T}} \mathrm{D}_{\frac{1}{\beta}} \mathbf{b}^{\prime}  \tag{4.1-17}\\
& =C_{0}-\frac{1}{2} \mathbf{b}^{\prime \mathrm{T}} \mathrm{D}_{\frac{1}{\beta}} \mathbf{b}^{\prime} .
\end{align*}
$$

The strategy is to now choose a value, $\bar{C}_{\text {aim }}$, as the constraint statistic value to be aimed at by this iteration. Choosing $\bar{C}_{\text {aim }}$ less than $\bar{C}_{\min }$ would be foolish. Even choosing $\bar{C}_{\text {aim }}=\bar{C}_{\min }$ is not ideal as this would leave no room for maximising entropy. Skilling and Bryan (1984) use the value,

$$
\begin{equation*}
\bar{C}_{\text {aim }}=\frac{2}{3} \bar{C}_{\min }+\frac{1}{3} C_{0} \tag{4.1-18}
\end{equation*}
$$

that is, two-thirds of the way towards the minimum of $\bar{C}(\mathbf{x}) .{ }^{2}$
Within the subspace, there will be many points with the constraint statistic

[^19]value of $\bar{C}_{\text {aim }}$, but the one with maximum entropy must be found ${ }^{3}$. A new function is constructed to help find this point,
\[

$$
\begin{equation*}
\bar{H}(\mathbf{x})-\alpha \bar{S}(\mathbf{x})-\bar{C}(\mathbf{x}) \tag{4.1-19}
\end{equation*}
$$

\]

where $\alpha$ is a Lagrange multiplier. This function is to be maximised, so,

$$
\frac{\partial \bar{H}}{\partial \mathbf{x}}=\alpha \frac{\partial \bar{S}}{\partial \mathbf{x}}-\frac{\partial \bar{C}}{\partial \mathbf{x}}=\alpha\left(\mathbf{a}^{\prime}-\mathbf{x}\right)-\left(\mathbf{b}^{\prime}+\mathrm{D}_{\beta} \mathbf{x}\right)-0
$$

which gives,

$$
\begin{equation*}
\mathbf{x}-\mathrm{D}_{\frac{1}{\beta+\alpha}}\left(\alpha \mathbf{a}^{\prime}-\mathbf{b}^{\prime}\right) \tag{4.1-20}
\end{equation*}
$$

As $\alpha$ varies, this equation defines the maximum entropy trajectory through the subspace. Following this trajectory will always result in maximum entropy points. It is possible to search along this trajectory until the point with value $\bar{C}_{\text {aim }}$ is found. Figure 4.1-1 schematically demonstrates the significance of the points $\mathbf{x}$ within the subspace. When $\alpha=\infty$ in equation (4.1-20), $\mathbf{x - \mathbf { a } ^ { \prime }}$, and when $\alpha=0, \mathbf{x}--\mathbf{D}_{\frac{1}{\beta}} \mathbf{b}^{\prime}$. These vectors define the maximum of entropy and the minimum of the constraint statistic and represent the ends of the maximum entropy trajectory.

Skilling and Bryan (1984) implemented a simple " $\alpha$-chop" algorithm for finding the required point on the maximum entropy trajectory. This algorithm steps through $\alpha$ values until a value is found that results in a constraint statistic very close to $\bar{C}_{\text {aim }}$. This is stable because $\bar{C}$ varies monotonically along the maximum entropy

[^20]trajectory due to the quadratic approximation. Even so, the quadratic approximation will not be valid too far away from the origin of the subspace coordinates ${ }^{4}$. For this reason, a distance limit, $L$, is imposed. If $\bar{C}_{\text {aim }}$ cannot be achieved within the distance limit, then the intersection of the distance limit and the smallest $\bar{C}$ value on the maximum


Fig. 4.1-1 A schematical representation of the subspace showing the maximum entropy trajectory.
entropy trajectory is chosen. A more severe problem can occur (rarely) if the current model is poor. There may be no point on the maximum entropy trajectory within the distance limit. This case is handled (see Skilling and Bryan, 1984) by including an extra term to the function $\bar{H}(\mathbf{x})$ which includes the length of $\mathbf{x}$ and an extra Lagrange multiplier. The Lagrange multiplier is normally kept at a zero value and is only increased in the rare case of having no valid points within the distance limit.

Once the required point $\mathbf{x}$ has been chosen in the subspace, it can be mapped back to the full $n$-dimensional space. This is done using the mapping of equation (4.18) after suitable transformation to account for the simultaneous diagonalisation (see section 4.2). The projection of $\mathbf{x}$ into the full space gives the updated model for this iteration.

A very important parameter used for checking the performance of the maximum

4 This is the location of the current model. Recall equations (4.1-10) and (4.1-11).

NB. In this figure, the gradient of $S$ is pointing in the direction of steepest ascent whilst the gradient of C is pointing in the direction of steepest descent.
entropy inversion algorithm is the TEST parameter (see Skilling and Bryan, 1984).
Figure 4.1-2 shows that the gradients of the entropy surface and the constraint statistic surface must be parallel at the true maximum entropy point. The TEST parameter measures the degree of non-parallelism and is defined as,

$$
\begin{equation*}
\text { TEST }-\frac{1}{2}\left|\frac{\nabla S}{|\nabla S|}-\frac{\nabla C}{|\nabla C|}\right|^{2} \tag{4.1-21}
\end{equation*}
$$

It is important that this parameter remains small throughout the inversion. As will be seen in section 4.4, this parameter is useful for monitoring the stability of the inversion as well as the suitability of the quadratic approximations and distance limits.

FORTRAN code for applying the above procedure was supplied by Steve Brown (The University of Sydney). This code was modified as described in the following three sections. The final code consists of a group of subroutines driven by MEMNL\$8 (see the appendix).


Fig. 4.1-2 The solid curves represent contours of $S$. The dashed curve is the contour of $C$ with a value of $C_{\text {aim }}$. Maximum entropy points require parallel gradients.
entropy inversion algorithm is the TEST parameter (see Skilling and Bryan, 1984).
Figure 4.1-2 shows that the gradients of the entropy surface and the constraint statistic surface must be parallel at the true maximum entropy point. The TEST parameter measures the degree of non-parallelism and is defined as,

$$
\begin{equation*}
\text { TEST }=\frac{1}{2}\left|\frac{\nabla S}{|\nabla S|}-\frac{\nabla C}{|\nabla C|}\right|^{2} \tag{4.1-21}
\end{equation*}
$$

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Fig. 4.1-2 The solid curves represent contours of $S$. The dashed curve is the contour of $C$ with a value of $C_{\text {aim }}$.

NB. In this figure, the gradient of S is pointing in the direction of steepest ascent whilst the gradient of C is pointing in the direction of steepest descent.
4.2 Simultaneous diagonalisation within the subspace.

As mentioned in the previous section, Skilling and Bryan (1984) defined an entropy metric to assist with the large dynamic range problems that accompany inversions of astronomical data. Using this metric affects the theory of the simultaneous diagonalisation of matrices A and B (see section 4.1). It is not essential to develop this theory in terms of the entropy metric. The purpose of this section is to define precisely how this diagonalisation is to be performed.

Recall the subspace representations of the quadratic approximations (equations (4.1-12) and (4.1-13)),

$$
\begin{align*}
& \bar{S}(\mathbf{y})=S_{0}+\mathbf{y}^{\mathrm{T}} \mathbf{a}-\frac{1}{2} \mathrm{y}^{\mathrm{T}} \mathrm{~A} \mathbf{y}  \tag{4.2-1}\\
& \bar{C}(\mathbf{y})=C_{0}+\mathbf{y}^{\mathrm{T}} \mathbf{b}+\frac{1}{2} \mathbf{y}^{\mathrm{T}} \mathrm{~B} \mathbf{y}, \tag{4.2-2}
\end{align*}
$$

where,

$$
\begin{aligned}
& \mathbf{a}=\mathrm{E}^{\mathrm{T}} \nabla S_{0}, \quad \mathrm{~A}-\mathrm{E}^{\mathrm{T}}\left(-\nabla \nabla S_{0}\right) \mathrm{E}, \\
& \mathbf{b}=\mathrm{E}^{\mathrm{T}} \nabla C_{0}, \quad \mathrm{~B}=\mathrm{E}^{\mathrm{T}}\left(\nabla \nabla C_{0}\right) \mathrm{E} .
\end{aligned}
$$

Here $\mathbf{a}$ and $\mathbf{b}$ are ( $r \times 1$ )-vectors and $\mathbf{A}, \mathrm{B}$ are ( $\mathbf{r} \times \mathrm{r})$-matrices. Now, $-\nabla \nabla S_{0}$ is positive definite ${ }^{1}$ which makes A real, symmetric and positive definite, and B real and symmetric.
${ }^{1}$ The second derivative of $S$ (equation 3.2-2) is $\frac{\partial^{2} S}{\partial q_{i} \partial q_{j}}=\frac{-\delta_{i j}}{q_{j}}$. The entries of $-\nabla \nabla S_{0}$ are all positive.

The aim is to find a transformation $H$ which simultaneously diagonalises A and B. Define the transformation as,

$$
\begin{equation*}
y-H x \tag{4.2-3}
\end{equation*}
$$

so that equations (4.2-1) and (4.2-2) become,

$$
\begin{align*}
& \bar{S}(\mathbf{x})-S_{0}+\mathbf{x}^{\mathrm{T}}\left(\mathrm{H}^{\mathrm{T}} \mathbf{a}\right)-\frac{1}{2} \mathbf{x}^{\mathrm{T}}\left(\mathrm{H}^{\mathrm{T}} \mathrm{AH}\right) \mathbf{x}  \tag{4.2-4}\\
& \bar{C}(\mathbf{x})=C_{0}+\mathbf{x}^{\mathrm{T}}\left(\mathrm{H}^{\mathrm{T}} \mathbf{b}\right)+\frac{1}{2} \mathbf{x}^{\mathrm{T}}\left(\mathrm{H}^{\mathrm{T}} \mathrm{BH}\right) \mathbf{x}, \tag{4.2-5}
\end{align*}
$$

It is now possible to invoke a well known theorem which states: if $A$ is real, symmetric and positive definite and $\mathbf{B}$ is real and symmetric then there exists a transformation, H , such that $\mathrm{H}^{\mathrm{T}} \mathrm{AH}=\mathrm{I},{ }^{2}$ and $\mathrm{H}^{\mathrm{T}} \mathrm{BH}$ is diagonal.

In particular $H$ can be constructed from the eigenvectors of A and B. If $\mathrm{U}=\left[\mathbf{u}_{1}, \mathbf{u}_{2}, \ldots, \mathbf{u}_{\mathrm{r}}\right]$ is the $(r \mathrm{x} r)$-matrix of normalised eigenvectors of A , then $U^{T} U=I$ and $U^{T} A U=D_{\alpha}$ where $D_{\alpha}$ is the diagonal matrix consisting of the eigenvalues of $A,\left(\alpha_{1}, \alpha_{2}, \ldots, \alpha_{r}\right)$. All the eigenvalues are positive since $A$ is positive definite. Now define,

$$
\begin{equation*}
\mathrm{G}=\mathrm{UD} \frac{1}{\sqrt{\alpha}}=\left[\frac{\mathbf{u}_{1}}{\sqrt{\alpha_{1}}}, \frac{\mathbf{u}_{2}}{\sqrt{\alpha_{2}}}, \ldots, \frac{\mathbf{u}_{\mathbf{r}}}{\sqrt{\alpha_{r}}}\right] \tag{4.2-6}
\end{equation*}
$$

so that,

$$
\begin{equation*}
G^{T} G-D_{\frac{1}{\sqrt{\alpha}}} U^{T} U D_{\frac{1}{\sqrt{\alpha}}}-D_{\frac{1}{\sqrt{\alpha}}} D_{\frac{1}{\sqrt{\alpha}}}=D_{\frac{1}{\alpha}} \tag{4.2-7}
\end{equation*}
$$

and,

[^21]\[

$$
\begin{equation*}
\mathrm{G}^{\mathrm{T}} \mathrm{AG}-\mathrm{D}_{\frac{1}{\sqrt{\alpha}}}\left(\mathrm{U}^{\mathrm{T}} \mathrm{AU}\right) \mathrm{D}_{\frac{1}{\sqrt{\alpha}}}-\mathrm{D}_{\frac{1}{\sqrt{\alpha}}} \mathrm{D}_{\alpha} \mathrm{D}_{\frac{1}{\sqrt{\alpha}}}=\mathrm{I} . \tag{4.2-8}
\end{equation*}
$$

\]

Also let $B^{\prime}=G^{T} B G$ which will remain real and symmetric. This $B^{\prime}$ can be diagonalised in the same way as above, with the use of the matrix of normalised eigenvectors of $\mathbf{B}^{\prime}$, namely V . That is,

$$
\begin{equation*}
V^{T} B^{\prime} V-D_{\beta} . \tag{4.2-9}
\end{equation*}
$$

Notice that the eigenvalues of $\mathbf{B}^{\prime}$ are $\left(\beta_{1}, \beta_{2}, \ldots, \beta_{\mathrm{r}}\right)$ and $\mathrm{V}^{\mathrm{T}} \mathrm{V}-\mathrm{I}$. Construct,

$$
\begin{equation*}
\mathrm{H}-\mathrm{GV}-\mathrm{UD}_{\frac{1}{\sqrt{\alpha}}} \mathrm{~V} \text {, } \tag{4.2-10}
\end{equation*}
$$

and consider,

$$
H^{T} A H-V^{T}\left(G^{T} A G\right) V-V^{T} I V-V^{T} V-I
$$

and,

$$
H^{\mathrm{T}} \mathrm{BH}-\mathrm{V}^{\mathrm{T}}\left(\mathrm{G}^{\mathrm{T}} \mathrm{BG}\right) \mathrm{V}-\mathrm{V}^{\mathrm{T}} \mathrm{~B}^{\prime} \mathrm{V}-\mathrm{D}_{\beta} .
$$

Therefore, the transformation H as defined by equation (4.2-10) is the required transformation for the simultaneous diagonalisation of A and B within the subspace. Application of this operator leads directly to equations (4.1-14) and (4.1-15).

[^22]
### 4.3 Including stages of decreasing smoothing and weighted least squares.

The rationale and purpose of using stages of decreasing smoothing within the inversion process was described in section 2.5 . Actually including this technique imposes a significant computational burden. Care has to be taken to ensure that the code is as efficient as possible.

Recall the derivatives of the constraint statistic from section $2.5^{1}$,

$$
\begin{equation*}
\nabla C-2 C_{\mathrm{m}} \nabla \mathrm{X}_{\mathrm{er}}^{\mathrm{T}} \mathbf{X}_{\mathrm{err}}+2\left(\mathrm{v}-\mathbf{v}_{0}\right), \tag{4.3-1}
\end{equation*}
$$

and,

$$
\begin{equation*}
\nabla \nabla C=2 C_{m} \nabla \mathrm{X}_{\mathrm{erf}}^{\mathrm{T}} \nabla \mathrm{X}_{\mathrm{err}} \mathrm{C}_{\mathrm{m}}+2 \mathrm{C}_{\mathrm{m}}, \tag{4.3-2}
\end{equation*}
$$

where $\mathrm{C}_{\mathrm{m}}$ is the model covariance matrix which is constructed with Gaussian functions for the smoothing (see section 2.5). Without the stages of smoothing technique, these equations would be,

$$
\nabla C-2 \nabla \mathrm{X}_{\mathrm{err}}^{\mathrm{T}} \mathbf{X}_{\mathrm{err}}
$$

and,

$$
\nabla \nabla C-2 \nabla \mathrm{X}_{\mathrm{er}}^{\mathrm{T}} \nabla \mathrm{X}_{\mathrm{efr}}
$$

The inclusion of the model covariance matrix, $\mathrm{C}_{\mathrm{m}}$, in these expressions is significant as $\mathrm{C}_{\mathrm{m}}$ is a ( $n \times n$ )-matrix where ' $n$ ' is the number of discrete parameters in the model. Whenever $\nabla C$ or $\nabla V C$ is required, this large matrix must be accessed and multiplied into the calculation. For the case of $\nabla V C$ this multiplication is two-fold.

Fortunately $\mathrm{C}_{\mathrm{m}}$ is symmetric and many of the entries are not independent. The

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Recall the derivatives of the constraint statistic from section $2.5,{ }^{1}$

$$
\begin{equation*}
\nabla C-2 C_{m} \nabla X_{e r r}^{\mathrm{T}} \mathbf{X}_{\mathrm{err}}+2\left(\mathrm{v}-\mathrm{v}_{\mathrm{o}}\right), \tag{4.3-1}
\end{equation*}
$$

and,

$$
\begin{equation*}
\nabla \nabla C-2 C_{m} \nabla \mathrm{X}_{\mathrm{erf}}^{\mathrm{T}} \nabla \mathrm{X}_{\mathrm{err}} \mathrm{C}_{\mathrm{m}}+2 \mathrm{C}_{\mathrm{m}}, \tag{4.3-2}
\end{equation*}
$$

where $\mathrm{C}_{\mathrm{m}}$ is the model covariance matrix which is constructed with Gaussian functions for the smoothing (see section 2.5). Without the stages of smoothing technique, these equations would be,

$$
\nabla C-2 \nabla \mathrm{X}_{\mathrm{err}}^{\mathrm{T}} \mathbf{X}_{\mathrm{err}}
$$

and,

$$
\nabla \nabla C-2 \nabla \mathrm{x}_{\mathrm{er}}^{\mathrm{T}} \nabla \mathrm{X}_{\mathrm{erf}} .
$$

The inclusion of the model covariance matrix, $\mathrm{C}_{\mathrm{m}}$, in these expressions is significant as $\mathrm{C}_{\mathrm{m}}$ is a $(n \mathrm{x} n$ )-matrix where ' $n$ ' is the number of discrete parameters in the model. Whenever $\nabla C$ or $\nabla \nabla C$ is required, this large matrix must be accessed and multiplied
${ }^{1}$ The second term on the right of this equation disappears when $\nabla C$ is evaluated at the current model, $\mathrm{v}_{0}$, at each step of the iterative inversion.
numerical value of the entry $\left(C_{m}\right)_{i j}$ depends on the physical distance separating the grid nodes represented by $i$ and $j$ (see equation (2.5-7)). For a discrete two-dimensional grid with a total of $n$ nodes, there are $n$ different physical separations possible. The ( $n$ $\mathrm{x} n$ )-matrix $\mathrm{C}_{\mathrm{m}}$ can therefore be completely stored as a vector of $n$ numbers.

Multiplying an ( $n \times n$ )-matrix by a ( $n \times n$ )-matrix requires $n^{3}$ multiplications, however, multiplying an ( $n \times n$ )-matrix by an ( $n \times 1$ )-vector requires only $n^{2}$ multiplications. As $n$ becomes large, straight matrix calculations are prohibitive. Fortunately the inversion process described in sections 4.1 and 4.2 can be performed without the necessity of such matrix computations. This means that every time $\nabla \nabla C$ is required in a computation, it is recalculated, but since these calculations are all matrixvector products, the multiplications required total much less than $n^{3}$.

The FORTRAN code that performs the maximum entropy inversion is a group of subroutines driven by a program MEMNL\$8 (see the appendix). The original code for this inversion was supplied by Steve Brown. Major modifications have been made to the code to tailor it to seismic reflection tomography and to include the model covariance matrix.

### 4.4 Automatic inversion control.

As described in section 4.1, the inversion process is based on quadratic approximation of both $C$ (the constraint statistic) and $S$ (entropy) within a 4 or 6-dimensional subspace. In general, the $C$ surface will be more non-linear than a quadratic function. It is important that movements within the model space are kept within the region of acceptable quadratic approximation. The distance limit (see section 4.1), $L$, is used precisely for this purpose. However, it is not a simple matter to determine the limits of acceptable quadratic approximation.

Another problem for the control of the inversion is suitable criteria for stopping a smoothing stage (see section 2.5 ) and beginning the next stage with less smoothing. A smoothing stage is to be terminated when no further significant $C$ reductions are to be achieved. It is not simple to determine when this occurs. It must be remembered that $C$ is non-linear and that it is possible for the rate of $C$ decrease to vary significantly. The inversion stage may be prematurely terminated at a point where a couple of iteration produce little decrease, even though some significant decreases could be made in later iterations. Another possibility is that the distance limit can be so small that successive iterations result in little $C$ reduction, even though sizeable, stable reductions may be achieved if the distance limit is relaxed. Without more complete knowledge of the non-linear $C$ surface, these problems may be difficult to overcome.

Fortunately, some of the parameters available from the inversion process itself give some insight into the success and applicability of the current inversion iteration.

Parameters such as the recent history of $C$ reductions, the actual length of the iteration update compared to the distance limit, and the changing values of the TEST parameter (see section 4.1) are all very useful for specifying the requirements for suitable control of the inversion process. Using these parameters, the software can be coded to automatically adapt to the changing conditions that confront the iterative inversion.

The most important parameter that needs to be adaptive for a successful inversion is the distance limit, $L$ (see section 4.1). This limit must have the ability to either increase or decrease rapidly, when necessary. The structure of the C surface can change dramatically between iterations, and especially between stages with different smoothing. A successful distance limit in one stage may be inappropriate in the next stage. Generally, larger distances can be traversed safely when the smoothing is heavier.

The algorithm for the automatic, adaptive control of the inversion process was coded into the main driver routine, METOMOS6 (see the appendix). This algorithm is now outlined in point form.
a) The $C$ values of the most recent 5 iterations are recorded in an array. Let these be known as $C_{m e m(i)}$ where $i$ ranges from 1 to 5 , and $C_{m e m(5)}$ being the most recent value.
b) If any of $C_{m e m(2)}$ through $C_{m e m(5)}$ is more than $2 \%$ lower than $C_{m e m(1)}$, then set a flag to signify that sufficient $C$ reduction has been achieved in recent iterations. This value of $2 \%$ is arbitrary but it seems to work well in practice. If less than 5 iterations have been performed, this flag is always set.
c) Now check for one of the two bad conditions.
(1) TEST $>0.2$ and TEST $>$ TESTOLD. Here 0.2 is the (somewhat arbitrary ${ }^{1}$ ) choice of the target TEST value and TESTOLD is the TEST value from the previous iteration. This condition implies that the current image is too far from the maximum entropy image and that it is moving further away.
(2) Two or more of the last 5 iterations have resulted in increases of $C$. This condition implies that the distance limit maybe too large and the quadratic approximation is only valid over a smaller range.

Given either of these conditions, the distance limit is halved, the counter UCOUNT is incremented, all $C_{m e m(i)}$ are set to $C_{m e m(5)}$ and the execution is sent back for the next iteration. The UCOUNT parameter records the number of times the inversion has been detected to be unstable. Setting all $C_{m e m(i)}$ to the value of $C_{m e m(5)}$ gives the next iteration, with a smaller distance limit, a clean start. Previous problems are therefore not allowed to plague the following iterations.

If either condition (1) or (2) is satisfied and the distance limit is already very small, and TEST is less than 0.2 , then the stage is ended as further distance limit reduction will be pointless.
d) Next, if the $2 \%$ flag from b) has not been set and TEST is less than 0.2 , then assume that as much $C$ reduction as possible has been achieved and terminate the current stage. Termination will proceed unless the current update distance is more than $90 \%$ of the distance limit and UCOUNT (the instability count) is less than 10 . In this case, the inversion may well be in a stable situation but simply stifled by a small distance limit. The action here is to increase the distance limit by $3 / 2$ and go onto the

[^23]next iteration.
The UCOUNT parameter is used to stop generally unstable inversions from getting any more distance limit increases. If the inversion has appeared temporarily unstable more than 9 times already (ie. UCOUNT $>10$ ) then it is probably not worth giving it any more chances with a larger distance limit.
e) The last check is for a stable inversion that may benefit from a slightly larger distance limit, hopefully improving the convergence rate. Here, if there were no $C$ increases in the last 5 iterations, and the current update distance is more than $90 \%$ of the distance limit, and UCOUNT is less than 10, then increase the distance limit by $5 / 4$. The increases suggested here and in d) have been chosen with the help of practical experimentation.
f) If no action has been enforced by any of the above conditions, simply begin the next iteration without any changes.

The objective of this automatic adaptive inversion control is two-fold: (i) to try and avoid early terminations when considerable reductions were still possible, and (ii) to avoid late terminations where large computational expense is wasted for very little gain. It is difficult to get the balance right but practical results show that the above algorithm works well in most cases.

In simple terms, the procedure described above treats $C$ increases and TEST increases (above 0.2) as indicators of possible instability. These are indications that the extent of valid quadratic approximation may be less than the distance limit. If these conditions persist, the distance limit is reduced. On the other hand, the distance limit can be increased if everything appears stable and the current iteration update is being stopped at the distance limit. In this case, the quadratic approximation may be valid
beyond the distance limit and faster convergence may be achieved with a larger distance limit. The range of validity of the quadratic approximation to $C$ will vary as the local topology of $C$ varies. The automatically controlled inversion, as described above, is designed to adapt the distance limit to the changing conditions encountered as the iterations proceed.

## §5 SYNTHETIC AND REAL DATA EXAMPLES

### 5.1 Tests with square-anomaly model

It is important to have a simple, yet significant, problem for the evaluation of the algorithm described in the earlier sections. Simplicity is needed to ensure that the algorithms success can be clearly measured. For this purpose, the model shown in Figure 5.1-1 has been developed (this model was also used in section 2.4). This model consists of a 2 km by 2 km subsurface area with a background velocity consisting of a simple linear increase with depth of $0.75 \mathrm{~m} / \mathrm{s}$ per second. The velocity at the surface is $1500 \mathrm{~m} / \mathrm{s}$. Within this background velocity field, a $400 \times 400$ metre square anomaly with a constant velocity of $2250 \mathrm{~m} / \mathrm{s}$ is centred. This anomaly velocity has a velocity increase for the top half and a decrease for the bottom half. The velocity of the background and the anomaly are identical at a depth of 1000 metres. Figure 5.1-2 shows the model with the velocities displayed as contours.

Traveltime data were computed for this model by tracing rays from regular source and receiver locations and a range of surface angles. A schematical demonstration of the types of rays that


Fig. 5.1-1 The velocities and dimensions of the model to be studied in section 5.1.


Fig. 5.1-2 The model of Figure 5.1-1 displayed as a contour plot. The contour interval is $25 \mathrm{~m} / \mathrm{s}$.


Fig. 5.1-3 The types of rays traced to obtain the traveltime data. The true paths bend where the velocities vary.
${ }^{1}$ The RMS $X_{e r r}$ value for the reconstruction in Figure 5.1-4 is 3.46 metres whereas for the reconstruction of Figure 5.1-5 it is 9.95 metres. There is an apparent paradox here, the better solution has a larger error. This comes about since the 184 traveltime data set was unable to identify the model but was still able to vary the velocities to match the traveltimes quite well. Consider a data set consisting of a single ray. Clearly the velocities can be varied such the residual error will be zero, yet the solution will not be satisfactory.
were traced is shown in Figure 5.1-3 (Figure 2.4-4 shows some of the actual raypaths). The objective of these ray locations was to provide a uniform range of source/receiver offsets and penetration depths.

Initially only 184 rays were traced for the traveltime data (the ones actually displayed in Figure 2.4-4). An inversion of these data can be seen in Figure 5.1-4. The starting model for the inversion consists of the linear background field only. This inversion used finite-difference derivative estimates and no constraint statistic smoothing. The value of the constraint statistic was reduced from about 140,000 to just 2197 , but the anomaly has not been resolved. Traveltime data was then obtained for 1312 raypaths (with surface locations spaced uniformly across the extent of the model) and an identical inversion performed. The final image of Figure 5.1-5 shows that the extra raypaths, implying more complete angular coverage, have resulted in a reasonable image of the anomaly. The 184 traveltime data set did not contain enough information for the inversion to recognise the anomaly. ${ }^{1}$

Section 2.4 discussed the theory behind the semi-analytic derivative estimates. This was developed because finite-difference derivative estimates were becoming very time consuming when 1,312 rays were used. Inversion results with semi-analytic derivative estimates were shown in section 2.4 and are reproduced in Figure 5.1-6. Surprisingly, these results were an improvement over the equivalent results with finite-difference derivative estimates. Possible reasons for this improvement were discussed in section 2.4. Further improvements of the final image were obtained when stages of decreasing smoothing were included in the inversion (see Figure 5.1-7). The stages applied here had Gaussian half widths (see section 2.5) of $125,67,33$ and 0 metres. This inversion has done a good job of imaging the anomaly, however, a constant velocity of $2250 \mathrm{~m} / \mathrm{s}$ within
were traced is shown in Figure 5.1-3 (Figure 2.4-4 shows some of the actual raypaths). The objective of these ray locations was to provide a uniform range of source/receiver offsets and penetration depths.

Initially only 184 rays were traced for the traveltime data (the ones actually displayed in Figure 2.4-4). An inversion of these data can be seen in Figure 5.1-4. The starting model for the inversion consists of the linear background field only. This inversion used finite-difference derivative estimates and no constraint statistic smoothing. The value of the constraint statistic was reduced from about 140,000 to just 2197 , but the anomaly has not been resolved. Traveltime data was then obtained for 1312 raypaths (with surface locations spaced uniformly across the extent of the model) and an identical inversion performed. The final image of Figure 5.1-5 shows that the extra raypaths, implying more complete angular coverage, have resulted in a reasonable image of the anomaly. The 184 traveltime data set did not contain enough information for the inversion to recognise the anomaly. ${ }^{1}$

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Fig. 5.1-4 Inversion of 184 traveltimes. Finite difference derivative estimates and no smoothing. Final C value of 2197.


Fig. 5.1-5 An identical inversion to that of Fig 5.1-4 except that 1312 traveltimes were used. Final C value is approximately 130,000 .


Fig. 5.1-6 An identical inversion to that of Fig. 5.1-3 except that semi-analytic derivative estimates were used. The final C value is approximately 50,000 .
the anomaly has not resulted. The synthetic traveltime data is noise free, so the less than perfect image reconstruction signifies imperfections in the specifications of the problem. Inherent ambiguities exist that are not associated with noise.

All of the above inversions were initialised with the correct background velocity field as the starting model. In general, this background velocity field would not be known accurately. The image of Figure 5.1-8 is the result of an inversion that used an incorrect background field as the starting velocity model. The linear velocity increase of the background was set to $0.825 \mathrm{~m} / \mathrm{s}$ per metre instead of $0.75 \mathrm{~m} / \mathrm{s}$ per metre. This inversion did not use stages of decreasing smoothing. At first glance the anomaly appears to be imaged just as well as when the correct background starting model was used (Figure 5.16). However, closer inspection reveals that the anomaly has been imaged with velocities that are too fast and at a depth that is too deep. The inversion has been trapped by a local


Fig. 5.1-7 Improved inversion results when stages of decreasing smoothing are added to the inversion used if Fig. 5.1-6. The final C value is approximately 43,000.


Fig. 5.1-8 The final image of an identical inversion to that of Fig. 5.1-6, ecept that the background linear gradient was $0.825 \mathrm{~m} / \mathrm{s}$ per metre. The final C value is approximately 59,000.
minimum.
Incorporation of stages of decreasing smoothing into the inversion (see section 2.5) should help to avoid local minima. Figures 5.1-9 through 5.1-12 record the progress of an inversion with stages of decreasing smoothing. Figure $5.1-10$ shows that the velocity field has been slowed by the first stage where the smoothers had half-widths of 2000 metres; this is the type of adjustment needed to correct the erroneous starting velocity model. The second stage, with smoothing half-widths of 1000 metres, continues the correction (Figure 5.1-11). These first two stages have approximately corrected the errors in the starting velocity model. The final stage of the inversion, without smoothing, has imaged the anomaly with correct velocities and depths (Figure 5.1-12). The local minimum that trapped the inversion of Figure 5.1-8 has been avoided.

Even though the synthetic data used in the above inversion are noise free, and computed using the same ray tracing as used in the inversion, the best image is still not exact and the final C value is significantly greater than zero. Either the inversion is still being trapped by a local minimum or it has encountered a broad plateau that has slowed convergence enough to bring the algorithm to a stop (see section 4.4). The usefulness of reflection tomography is seriously questioned by this observation.

Consider the circumstances of Figure 5.1-13. A ray is traced across the boundary between two layers. With the currently assigned velocities, a significant $X_{e r r}$ value results. The actual change required is a simple increase of $v_{2}$ (see Figure 5.1-13). However, it is clear that any increase in the velocity contrast at the boundary will bend the two halves of the ray towards each other and reduce $X_{e r r}$. The velocities near the ray's intersection with the boundary become very sensitive. Decreases of $v_{1}$ just above the boundary are


Fig. 5.1-9 A contour plot of the erroneous background starting model. The linear increase is $0.825 \mathrm{~m} / \mathrm{s}$ per metre.


Fig. 5.1-10 The image after the first smoothing stage with a half width 2000 metres. The final C calue is approximately $1,700,000$.


Fig. 5.1-11 The image after the second smoothing stage with a half-width of 1000 metres. The final C value is approximately $1,000,000$.


Fig. 5.1-12 The final image after the stage without any smoothing. C value is approximately 49,000 .
just as effective at reducing $X_{e r r}$ as are increases in $v_{2}$ just below the boundary. If enough rays of sufficiently varying geometry are not included in the inversion, there is a real possibility for low velocity zones to appear above the boundary. Such erroneous updates may allow the inversion to drift toward local minima.

The sharp boundaries of the square


Fig. 5.1-13 Extra contrast at the boundary of the layers can remedy the condition of non-zero Xerr. anomaly in the synthetic model described in this section present similar problems as the layer boundary in Figure 5.1-13. Low velocity zones are apparent near the upper corners of the anomaly in Figure 5.1-12. Fast velocity zones are also apparent near the lower corners. Without introducing more traveltimes from new ray geometries, one way to help the inversion find the global minimum is to improve the starting model. With less modification necessary, these erroneous velocity zones will be less likely to appear. A starting model was constructed with the velocities within the anomalous region equal to the average of the background and true anomaly velocities at each depth. Effectively half of the anomaly has been placed in the starting model. This starting velocity model was also used as the prior information model (see section 3) in the inversion with the final image shown in Figure 5.1-14. It is not surprising that this image is the best result so far. There is much less evidence of erroneous velocity zones around the anomaly. The final constraint statistic, C, value is now approximately 20,000 instead of 49,000 as in the image of Figure 5.1-12. This
confirms that the previous inversion was trapped by a local minimum quite near the global minimum. A better starting model has avoided this local minimum. A greater multiplicity and variety of ray paths may also avoid problems with such local minima.

The above example of superior results with improved starting models suggests possible benefits from combining geological interpretation with reflection tomography. After an initial tomographic inversion the image can be geologically interpreted and a new simple starting model can be constructed. Such a model has the potential to improve the final results. This is similar to the expected effect of the stages of decreasing smoothing. However, the smoothing approach will inherently struggle with sharp discontinuities that are not present in the starting model. A combined and complementary use of geological interpretation and smoothing may be required for optimum results.


Fig. 5.1-14 The inverted image achieved when the starting and prior information models consisted of 50 percent of the anomalous variation. The final C value here is approximately 20,000 .

### 5.2 Tests with fault model

The model created for the tests described in this section is a realistic layered model containing a single, high angle fault. This model was instructive in the optimum way to specify the starting velocity model. Useful and accurate results were only obtained once a suitable starting model was established.

The interval velocities and geometry of this fault model are shown in Figure 5.2-1. The model consists of three simple layers containing a normal fault with 100 metres of throw. The layer velocities are of realistic magnitude. Eight hundred rays with regular geometries (similar to those in section 5.1) were traced through this model. These rays produced the traveltime data used for the experimental inversions.

The starting model for the initial inversion of this synthetic data consisted of three layers without any hint of the fault. The layer velocities were the true velocities and the depths of the layer boundaries were the correct ones for the left side of the model. The final inversion image resulting from this starting model is disappointing (see Figure 5.2-2). The fault locations have barely been imaged and the depth of the layer boundaries on the right were unchanged. It appears that sharp discontinuities will tend not to be changed by a somewhat underdetermined image


Fig. 5.2-1 The interval velocities and geometry of the fault model.
reconstruction. The main reason for this is that the required changes are very large. The sharp layer boundary of the starting model has been placed at the wrong depth for the right side of the model. Some velocities have to be changed from 2000 metres per second to 2400 metres per second, an increase of $20 \%$. Sharp discontinuities suggest substantial prior knowledge about the velocity structure. One must be careful to ensure that unwarranted information is not expressed in the prior information model.

Consider an inversion where the starting velocity model is a simple linear function with depth. The prior information model (generally the starting model) does not express any knowledge of the layer boundaries in this case. The image of Figure 5.2-3 is the result of an inversion with such a starting (and prior information) model. The existence of a fault is clearly revealed - at least for the shallowest layer boundary. However, the linear starting model has not allowed precise imaging of the layer boundaries away from the fault ${ }^{1}$.

In reality, the existence of layered geology would be known from nearby well logs or even CDP stacking velocities. The precise depth of the layer boundaries will not be known and the fault may not be clear, but the layers would be observed. The prior information model should exhibit the layered geology but be rather non-committal with respect to the depths of the boundaries. Such an approach leads to an initial model with correct layer velocities but with a 100 metre zone of linear change from one velocity to the next. The results of an inversion with such a starting model can be seen in Figure 5.2-4. The fault break has been well imaged, especially for the shallower layer boundary. The layer boundary to the right of the fault has moved upwards - to the left of the fault

[^24]

Fig. 5.2-2 The final inversion image when the starting model consisted of three simple layers with sharp boundaries.


Fig. 5.2-3 The final image achieved when the starting and prior information models are simple linear increases with depth.


Fig. 5.2-4 The inversion image obtained when the starting model changes the velocities linearly over a 100 metre zone at the layer boundary.
it has moved downwards.
A closer look at this inversion reveals some apparent problems. The graph of the reduction of the constraint statistic (Figure 5.2-5) and change of the TEST parameter (Figure 5.2-6) shows disturbing oscillations and persistently high TEST values. When the constraint statistic increases, either the quadratic approximation within the subspace (see section 4) is poor or the distance moved within the subspace is large enough to be beyond the limit of reasonable approximation. A close investigation of this problem revealed that the distance limit in the inversion was allowed to be too large. The minimum of the quadratic approximation was far from the current location in image space, but the approximation is only valid for comparatively smaller distances. The oscillations observed in Figures 5.2-5 and 5.2-6 are signals of such approximation problems. The inversion code can be modified such that the distance limit is halved when oscillations are detected


Fig. 5.2-5 The decrease of the constraint statistic value during the inversion resulting in the image of Figure 5.2-4.


Fig. 5.2-6 The value of the TEST parameter during the inversion resulting in the image of Figure 5.2-4.


Fig. 5.2-7 The image obtained when the inversion reduces the distance limit upon detecting oscillations.
(this was discussed in section 4). Results of an inversion with this reducing distance limit can be seen in Figures 5.2-7, 5.2-8 and 5.2-9. Notice that the constraint statistic began to oscillate near the 55th iteration but then the resulting distance limit reductions produced a stable convergence by the 71st iteration. The TEST parameter also increases rapidly when the oscillations appeared (see Figure 5.2-9) but decreased rapidly when the distance limit was reduced. Oscillations of the constraint statistic and increases of the TEST parameter, are useful indicators of deficiencies in the local quadratic approximation, and form the basis of the automatic inversion control described in section 4.

These last two inversion results (Figures 5.2-4 and 5.2-7) effectively provide a comparison of results with and without entropy. The final TEST parameter values in Figures $5.2-6$ and $5.2-9$ show that the image of Figure $5.2-4$ is far from the maximum entropy image whereas Figure 5.2-7 is a maximum entropy image. The final constraint


Fig. 5.2-8 The decrease of the constraint statistic for the inversion resulting in Figure 5.2-7.


Fig. 5.2-9 The values of the TEST parameter during the inversion resulting in Figure 5.2-7.
statistic values are very similar (Figures 5.2-5 and 5.2-8). The maximum entropy image (Figure 5.2-7) exhibits less structure than the image of Figure 5.2-4. This extra structure represents the unwarranted image components that have been rejected by maximising the entropy function (see section 3). A more striking comparison between maximum entropy and non maximum entropy images was presented in section 3.7.

### 5.3 Tests with a layered/anomaly model

A synthetic data set was prepared for a subsurface model that consisted of simple layered geology and an anomalous rectangular area of faster velocity. The interval velocities and depths of these features are sketched in Figure 5.3-1. The anomalous zone has a velocity of $2700 \mathrm{~m} / \mathrm{s}$ which is $20 \%$ above the velocity of the host layer. Random ray geometry was used to generate 800 traveltime data values. The source and receiver locations, as well as the surface intercept angles, were computed using random number generating software. This leads to a somewhat unrealistic data set but it should be one with a wide range of raypath geometries.

Using a simple linear velocity increase with depth as the starting model, the image of Figure 5.3-2 results (see section 2.5 for more details on this result). The constraint statistic values and the TEST parameter values for this inversion are shown in Figures 5.3-3 and 5.3-4. This result was obtained using the stages of decreasing smoothing as described in section 2.5. The inversion has clearly uncovered the presence of the anomaly without revealing the specifics of the anomaly or layers.

Without the assistance of stages of decreasing smoothing (see section 2.5 ), the inversion process is incapable of


Fig. 5.3-1 The interval velocities and depths of the layered/anomaly model.
imaging the anomaly (see Figure 5.3-5). If entropy is not used as a constraint (see section 3) an image with additional spurious structure results (see Figure 5.3-6 and compare to Figure 5.3-2). Notice the extra structure at the lower left of the anomaly. The top corners of the anomaly are possibly better imaged, but the maximum entropy results (Figure 5.3-2) indicate that this additional structure is unjustifiable given the data provided.

With real reflection tomography problems, the state of knowledge of the subsurface velocities will generally be more sophisticated than a simple increase with depth. Where definite layers exist, as in the current model, the interval velocities of the layers will normally be estimated from nearby wells or CDP stacking velocities. In addition, the velocity of some layers will be known quite accurately (for example, the water layer in the marine exploration case). It will assist the inversion if such well


Fig. 5.3-2 The inversion obtained from the 800 random raypaths.


Fig. 5.3-3 The constraint statistic values for the inversion of Figure 5.3-2


Fig. 5.3-4 The TEST parameter values for the inversion of Figure 5.3-2
known layers have their velocities held constant throughout the iterations. The starting model for the above results was a simple linear increase of velocity below the initial layer of $2000 \mathrm{~m} / \mathrm{s}$. If the velocity of this shallowest layer is not changed throughout the inversion, the image of Figure 5.3-7 results. The image of the anomaly is considerably sharper and the constant velocity layer around the anomaly is beginning to be properly imaged. Removing some of the velocities from the model parameters simplifies the inversion problem. However, if some velocities are fixed to incorrect values, the inversion may be prevented from achieving useful results.

Consider now a starting model consisting of the actual velocities of the layers, but without the presence of the anomaly. For the inversion of the data with this starting model, all of the velocities will be fixed except for those of the layer that contains the anomaly. The final image of this inversion is displayed in Figure 5.3-8. As expected, this image is superior to all earlier images.

There are many reasons why correct starting velocities can be modified during any given iteration of the inversion. These include noise, incorrect ray locations, ambiguity and a lack of ray coverage. It is generally assumed that such incorrect modifications will be insignificant when compared to the correct changes. However, removing the possibility of incorrect velocity updates in areas of relative certainty can help improve inversion results.

In real reflection tomography problems, the surface incidence angles will not be known exactly. Such circumstances can be evaluated using the current synthetic data if errors are added to the surface incidence angles of the rays. Figure 5.3-9 shows the image obtained after $10 \%$ gaussian noise has been added to the surface incidence angles. Figure 5.3-10 displays the results for $20 \%$ gaussian noise. With $10 \%$ noise the


Fig. 5.3-5 The image obtained from an inversion without stages of decreasing smoothing. (Final C value approximately 700,000 )


Fig. 5.3-6 The image obtained from an inversion without maximum entropy constraints. (Final C value approximately 200,000)


Fig. 5.3-7 The image obtained when the velocity of the first layer is held fixed. (Final C value approximately 200,000)


Fig. 5.3-8 The image obtained when the correct layer velocities are used as the starting model. All but the velocities for the anomalous layer were held constant at there correct values.


Fig. 5.3-9 The image obtained when $10 \%$ gaussian noise was added to the surface incidence angles. Compare to Figure 5.3-8


Fig. 5.3-10 The image obtained when $20 \%$ gaussian noise was added to the surface incidence angles. Compare to Figure 5.3-8
results are clearly degraded but still useful. With $20 \%$ noise, the results are of little use. The amount of surface incidence angle noise present in real data sets is clearly of significant importance.

### 5.4 Line A - Field data example

Real data examples present many problems that are not of concern with synthetic data examples. The data must be processed such that the automatic picking algorithm (see section 2.3) will pick reliable traveltimes and dips. Multiple energy and other noise must be sufficiently attenuated. Even with good data quality there will inevitably be some dip measurement errors. The examples in section 5.3 showed that these errors do not have to be much larger than $10 \%$ to cause serious problems. Added to these difficulties is the task of finding an adequate starting velocity model. Deep well control can make this generally troublesome task much easier.

Pre-stack depth migration provides a means for verifying subsurface velocity fields (see Al-Yahya, 1989). Iterative pre-stack depth migration can be used to define the subsurface interval velocity in the absence of well control. The real data example to be reviewed here, Line A, was the subject of exactly this type of iterative velocity evaluation. The final pre-stack depth migration is shown in Figure 5.4-1, and the corresponding velocity field is shown in Figure 5.4-2. This velocity field is then used as the starting velocity model needed for reflection tomography. The aim of the tomographic inversion is to improve this velocity model. An improved velocity model would produce a better pre-stack depth migrated image.

Traveltime and apparent dips were picked automatically using instantaneous frequency and wavenumber (see section 2.3). The many thousands of picks that resulted were decimated to approximately 2000 which were inverted using the entropy principle and stages of decreasing smoothing (half-widths of the gaussian smoothers


Fig. 5.4-1 The pre-stack depth migrated data - Line A. The velocity model used was obtained from iterations of pre-stack depth migration.


Fig. 5.4-2 The interval velocity model used for the pre-stack depth migration of Figure 5.4-1
were $400,200,100$ and 0 metres). The final velocity field is shown in Figure 5.4-3.
The inversion result (Figure 5.4-3) exhibits higher frequency structure than the starting model (Figure 5.4-2). It is not possible to accurately infer high frequency structure during velocity model building with iterations of pre-stack depth migration. The rate of spatial change of velocity that is possible with tomographic inversion is limited only by the chosen discrete node spacing and the quality of the data. Generally, the changes made appear realistic and the subsequent pre-stack depth migration results of Figure 5.4-4 demonstrate an improved image. Notice the stronger image of the shallower events and the changed attitude near the major faulting toward the left.

Accurate pre-stack depth migrations require accurate interval velocity fields and are characterised by a lack of residual moveout on the migrated common location


Fig. 5.4-3 The final image of the tomographic inversion of approximately 2000 traveltimes.


Fig. 5.4-4 The pre-stack depth migrated image obtained using the tomographically updated velocities of Figure 5.4-3
gathers (see Al-Yahya, 1989). Figures 5.4-5 and 5.4-6 display some migrated common location gathers using the velocity models of Figures 5.4-2 and 5.4-3. The data created with the tomographically derived velocities (Figure 5.4-6) exhibits generally flatter events, confirming that the updates made by the reflection tomography algorithm have been accurate.

It appears that reflection tomography can be useful in the role of updating previously estimated interval velocity fields. It has the potential to add higher frequency velocity variations than are obtainable by conventional means.


Fig. 5.4-5 Migrated common location gathers using the starting velocity model (Figure 5.4-2).


Time (sec)

Fig. 5.4-6 Migrated common location gathers obtained using the tomographically updated velocity field.

The example reviewed in this section was processed and evaluated in a similar way to Line A (described in the previous section). However, some further tests were performed on this line to explore a way of coping with the lack of precise knowledge regarding the depths of the layer boundaries.

The interval velocity model obtained from iterative pre-stack depth migration is shown in Figure 5.5-1, and the corresponding migrated image is in Figure 5.5-2. The processed gathers were picked using the complex attribute picker described in Section 2.3 . Approximately 2000 of these traveltimes were supplied to the reflection tomography software that inverted them using stages of decreasing smoothing and maximum entropy constraints. The final velocity field image is displayed in Figure 5.5-3. The corresponding pre-stack depth migration (Figure 5.5-4) shows improved imaging - especially at the edges of the major faulting (around times 0.7 to 1.0 seconds). As with Line A (see section 5.4), migrated common location gathers show generally less residual moveout (see Figures 5.5-5 and 5.5-6). It can be concluded that the tomographic velocity updates are accurate.

It was shown, using synthetic data in section 5.2 , that if the initial velocity model has a layer boundary specified at the incorrect depth, the inversion may be stifled. The required velocity changes may be too large to be accommodated. With both the current data and the data of Line A in the previous section, no well control was available so the depths to the layer boundaries were not accurately known. Only the two-way traveltimes of the reflections are known accurately. If the initial guess of


Fig. 5.5-1 The interval velocity model built from iterations of pre-stack depth migration.


Fig. 5.5-2 The pre-stack depth migrated data obtained using the velocity model of Figure 5.5-1


Fig. 5.5-3 The velocity field resulting from tomographic inversion using the velocities of Figure $5.5-1$ as the starting model.


Fig. 5.5-4 The pre-stack depth migrated image obtained using the tomographically updated velocities of Figure 5.5-3.


Fig. 5.5-5 Common location gathers - migrated using the starting velocity model (Figure 5.5-1).


Fig. 5.5-6 Common location gathers - migrated using the tomographically updated velocities (Figure 5.5-3). Generally less residual moveout here.
the interval velocities is wrong, then the implied depths will be wrong. Such potential problems could be avoided if the tomographic inversion updated a velocity field relative to time instead of depth.

Velocities specified relative to depth are mandatory for the raytracing component of the inversion process. However, it is possible to use a velocity field specified in time coordinates if it is temporarily mapped to depth for each iteration. Consider the following mapping strategy;

where $f_{1}$ is the first mapping from time to depth, and $f_{1}^{-1}$ is the inverse mapping (from
depth to time). The starting velocity model is specified in time. This is mapped to depth using the initial interval velocities and passed to the first iteration of the inversion. The interval velocities are changed in this iteration but the implied layer boundaries tend to remain at the same depths (see section 5.2). This implies that the times to the layer boundaries are changing even though these times are known quite accurately. An approximate way to change the depths instead of the times is to map the depth model back to time using the inverse of the previous forward mapping (ie. putting the layer boundaries back at the same times but with new velocities) and the creating a new forward mapping with the updated velocities. Such mapping, if performed between each iteration, will keep the times to the layer boundaries approximately constant and allow the implied depths to vary.

Such a time based inversion was run successfully on the data for Line B. The final velocity model and migrated data are displayed in Figures 5.5-7 and 5.5-8. Even though the velocity model (Figure 5.5-7) has some differences to that of Figure 5.5-3, the pre-stack depth migrated data is virtually unaffected. If the starting model had a larger velocity error, this technique may have greater effect.


Fig. 5.5-7 The velocity image resulting from an inversion based on a time referenced velocity model.


Fig. 5.5-8 The migrated image obtained with the velocity model of Figure 5.5-7.

The modified reflection tomography algorithm that has been constructed in the preceding sections, not only simplifies the traveltime interpretation but at the same time includes classes of reflected data that ${ }_{\wedge}$ aither difficult or impossible to include in the conventional tomographic inversion. Any event that has reflected or scattered from a single depth horizon can be picked and included in the inversion.

Diffraction events are a very important class of data that can be used in the current inversion algorithm. The conventional algorithm could not utilise such data since it is difficult to get two-point raytracing to trace a point reflector. These data will be very important in structurally complex areas. It is possible for diffracted arrivals to dominate in such areas. This feature also helps to alleviate the need for picking from approximately migrated traces (eg. Jacobs et al., 1992).

Fault plane reflections are another class of energy that is easily included in the tomographic inversion. These data can be very important because fault planes are often very steep and can provide some very high angle raypaths, enriching the angular coverage of the data. Extra angular coverage will improve the resolution of the reconstructed image and reduce the underdetermined nature of the inverse problem.

When the structure of the reflecting surfaces is significant, it is possible to get more than one reflection from the same surface to be recorded at any given receiver. These reflections are commonly called "multiple arrivals". Conventional reflection tomography can only use one of these arrivals - generally the first. Lailly et al., 1992, suggests that the first of the multiple arrivals is often the weakest and may not provide the
most reliable data. The tomographic inversion algorithm developed in this thesis naturally allows all of the multiple arrivals. Each arrival is distinguished by unique surface incidence angles (see section 2.2) and are all treated equally.

In this thesis it was assumed that the automatic traveltime picks (see section 2.3) will inevitably have some mispicks. Section 2.4 developed a data weighting scheme to help reduce the influence of these mispicks. However, these weights were based purely on the measured value of $X_{e r r(i)}$; this is not necessarily a reliable indicator of bad data. Some useful information will generally be suppressed as well. The following section looks at the possibility of using the entropy principle to assist in assigning more useful weights. This procedure has not been attempted so far in practice.

### 6.1 Further applications of entropy

In section 2.6 an arbitrary weighting scheme was developed in an attempt to limit the effect of erroneous data values on the inversion. The weights were based on the size of the individual $X_{e r r}$ values (see equation (2.6-4)). Large $X_{e r r}$ values may well be the result of valid velocity anomalies that need to be imaged. Even though these weights do assist the inverse procedure, a alternative weighting scheme may give better performance.

The discussion of section 4 indicates that the entropy principle can be used in problems of inference, and where more than one set of choices satisfies given constraints. It is quite possible that the entropy principle can play a role in assigning a better weighting scheme.

The critical information used by the inversion process is the set of partial derivatives of the constraint statistic with respect to each discrete velocity value, $\frac{\partial C}{\partial v_{j}}$ (see
section 3.4). This value is the sum of the $\frac{\partial X_{e r r(i)}}{\partial v_{j}}$ values for each ray influenced by this velocity, weighted by $2 X_{\text {err(i) }}$ (see equation (2.4-2)). It is suspected that a ray resulting from an erroneous data value will consistently have $\frac{\partial X_{e r r(i)}}{\partial v_{j}}$ values that differ markedly
from the mean value for that velocity node. Define the following function,

$$
\begin{equation*}
F_{i}=\sum_{j=1}^{N}\left(\frac{\partial \bar{X}_{e r r}}{\partial v_{j}}-\frac{\partial X_{e r r(i)}}{\partial v_{j}}\right)^{2} \tag{6.1-1}
\end{equation*}
$$

where there are $N$ discrete velocities values in the model and $\frac{\partial \bar{X}_{e r r}}{\partial v_{j}}$ represents the average derivative for each velocity node. It is now possible to define weights, $\omega_{i}$, based on these $F_{i}$ values rather than the $X_{e r r(i)}$ values. Further, define the weighted $F_{i}$ total,

$$
\begin{equation*}
\hat{F}-\sum_{i} \omega_{i} F_{i} \tag{6.1-2}
\end{equation*}
$$

Following the arguments of section 4 , the weights, $\omega_{i}$, can be specified with the aid of the entropy principle. With an entropy function defined as,

$$
\begin{equation*}
S_{\omega}-\sum_{i}\left(\omega_{i}-\omega_{i} \ln \omega_{i}\right) \tag{6.1-3}
\end{equation*}
$$

it is possible to find weights, $\omega_{i}$, that decrease $\hat{F}$ by some predefined amount and maximise $S_{\omega}$. These weights would discriminate against rays with Frechet derivatives that consistently deviated from the average derivative of the other rays. It is possible that such weights could damp the erroneous rays better than those based on the magnitude of $X_{\operatorname{err}(i)}$.

### 6.2 3D Applications

The reflection tomography algorithm described and tested in this thesis assumes that the seismic data are collected along a straight line, and that all reflected raypaths between the sources and receivers exist within the vertical plane passing through this line. The algorithm comprehends energy motion in two dimensions (2D) only. When the geological structures are complex, it becomes more difficult to approximate the 3D world with a 2D model. Three-dimensional seismic volumes are increasingly being collected because of this problem. It is important that the reflection tomography technique is capable of accurately imaging 3D problems.

Raytracing without reflector depth parameterisation (see section 2.2) in three dimensions requires surface intercept azimuths as well as angles, for both the source and receiver ends of the ray. With 3D seismic surveys collected on land, the intercept angle and azimuth can be determined directly if three-component receivers are used. Offshore, such three-component receivers are not practical. Conventional marine 3D seismic surveys are recorded as a closely spaced collection of parallel linear traverses (2D seismic lines). The direction of these parallel lines is called the 'inline' direction. It is sufficient to define inline and crossline components of the surface intercept angles, or inline angles and the ray azimuths, for the 3D raytracing (see Figure 6.2-1). The crossline components of the surface intercept angles are difficult to estimate. The inline components of the intercept angles can be estimated from common source and common


Fig. 6.2-1 Inline surface incidence angles ( $\theta$ 's) and raypath azimuths ( $\phi$ 's) suffice to begin 3D raytracing without pre-defined reflectors.
receiver gathers in the same way as for 2D data ${ }^{1}$ (see section 2.2). The crossline component may be estimated by gathering data from adjacent lines. If traces recorded by the same receiver on the cable are gathered and examined, across the lines, the crossline component of the intercept angle could be estimated. This approach is hindered by two problems; the crossline data spacing is usually much larger than the inline data spacing (often 50 metres versus 12.5 metres), and gathering crossline data traces from a 3D data volume can be very computer intensive.

It may be possible to perform successful 3D reflection tomographic inversions

[^25]

Fig. 6.2-2 An example of 3D raytracing in plan view. Once the depth has been reached where the total traveltime equals the measured time, $X_{\text {cross }}$ is the crossline component of $X_{\text {err }}$.
without having to estimate the crossline intercept angles from the data. If the ray azimuths (see Figure 6.2-1) are initially assumed to be zero, then raytracing would result in something like that shown in Figure 6.2-2. This figure is a plan view showing that when the depth is reached where the traced traveltime equals the measured traveltime, there will generally be a crossline component, $X_{\text {cross }}$, of the $X_{\text {err }}$ value. The source and receiver azimuths that were assumed to be zero can now be adjusted to remove the $X_{\text {cross }}$ component of the error. Such an azimuth adjustment can be made during every iteration. As long as the 3D starting velocity model is sufficiently close to
the actual velocity field, it is assumed that this iterative angle adjustment will converge to the true azimuth values.

An increasingly popular method of offshore seismic data collection consists of placing the receiver cable on the sea floor (known as 'ocean bottom cable' surveys). A major advantage of this method is that both a hydrophone and an onshore style velocity phone can be placed in the ocean bottom cable, allowing incidence angles and azimuths to be determined directly. This data collection scheme also detaches the sources from the receiver lines so that full angular coverage (although surface restricted) can be recorded. With ocean bottom cables, a 3D survey can be recorded as outlined in Figure 6.2-3. Such data collection should present better behaved tomographic inversions.


Fig. 6.2-3 A plan view of an ocean bottom cable 3D survey. The receiver cables are laid on the sea floor (solid lines) and the source boat is sailed (dashed lines) perpendicular to the receiver direction.

## §7 CONCLUSION

Many difficulties associated with conventional reflection tomography can be overcome. Raytracing without pre-defined reflector depths makes the traveltime interpretation simple rather than laborious - even automatic traveltime interpretation algorithms work satisfactorily. Maximising the entropy of the image plays a central role in stabilising the inversion of these traveltime data by ensuring that unjustifiable features are not reconstructed at any iteration. This is achieved by enforcing minimum commitment to the velocity of any given cell. Maximising entropy also helps to avoid ambiguity problems. Stages of decreasing smoothing applied to the constraint statistic of the inversion process is also useful for avoiding local minima.

An additional benefit offered by unrestricted traveltime data interpretation is that diffraction events, fault plane reflections and multiple arrivals can be picked and added into the inversion quite naturally. Any primary reflected energy can be used. Therefore, not only does the new raytracing approach allow many more data to be picked, it also allows for wider and fuller angular coverage. Synthetic data examples have indicated that wider angular coverage is more beneficial than simply having more traveltime data.

Synthetic data examples have demonstrated that stages of decreasing enstraint statistic smoothing can be extremely powerful. It can enable large anomalies (greater than $10 \%$ change from the starting model) to be imaged which went unnoticed by conventional inversions. When noise is added to the data used in the inversions, the ability to extract velocity anomalies is reduced. Tests with synthetic data showed that $10 \%$ surface angle noise is tolerable whereas $20 \%$ noise is not.

Using the entropy principle is of significant assistance to tomographic inversion. This principle prohibits velocity features from appearing that are not needed to satisfy the data. It should not be seen as a smoothness constraint - sharp changes can be imaged if required by the data. Maximising the entropy of the image during each iteration has a stabilising effect on the inversion as spurious structure is prevented from materialising. Also, the lack of unwarranted structure in the final solution considerably assists interpretation.

The entropy principle has a long history and has been used in many disciplines. It can be difficult to justify its use in tomographic image reconstruction. However, contingency table analysis and some very simple tomography problems lead to the required justification. Other popular inversion techniques use smoothness or minimum change constraints, and can lead to unwarranted, spurious structure in the solution image. The popular SIRT algorithm uses a minimum change (or minimum norm) constraint and synthetic examples have shown that more spurious structure exists on SIRT reconstructions compared to maximum entropy ones. Maximum entropy images are clearer and more interpretable.

Raytracing without fixed reflectors means that Fermat's principle cannot be invoked for Frechet derivative estimation. Initially it appeared that full re-raytracing would be needed to obtain the Frechet derivatives, but a 'semi-analytic' derivative estimation scheme proved to be accurate and efficient. The scheme requires raytracing through only a couple of discrete cells for each ray.

Raytracing without reflectors requires the apparent dips of the data on both common source and common receiver gathers. An algorithm based on the complex attributes of the traces has proven to be useful for this purpose, as well as being
computationally efficient.
The successes achieved with synthetic data sets are not as easily obtained with real data sets. The inherent noise problem with real data causes difficulties. Initially, the data must be adequately processed to attenuate multiple reflections as well as both coherent and random noise. Ideally, the data input to the automatic picking algorithm should consist of primary reflections only. Even so, the unavoidable noise in the data means that the starting velocity model for a tomographic inversion needs to be reasonably accurate. Fortunately, conventional interval velocity analysis methods can lead to a reasonably accurate starting model.

Two real data sets were considered and in both cases the starting velocity models were constructed from iterations of pre-stack depth migration. Pre-stack depth migration is also useful for evaluating the results of the tomographic inversion. After picking the data automatically and conducting the inversion with all the updates described above, the updated velocity model produced significant pre-stack depth migration improvements. Reflection tomography has a clear role in updating and improving the best velocity models obtained through conventional means.

Reflection tomography can, in principle, be easily extended to three-dimensions. Raytracing without reflector parameterisation requires ray azimuth estimation which may prove to be difficult. With conventional marine 3D recording, the azimuth can be obtained by considering adjacent lines. However, it may be possible to perform a successful 3D inversion without azimuth estimation. The azimuths could be iteratively obtained during the normal inversion iterations as long as the 3D starting model is adequate. Land 3D survey and marine ocean bottom cable 3D surveys offer the most promise in terms of 3D angular coverage and easy azimuth estimation.

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## APPENDIX

The FORTRAN subroutines listed in this appendix are described in the earlier text. The subroutines are listed in alphabetical order with the subroutine name posted at the top right of every page. Note that MEMNL\$8 is really a collection of a number of subroutines acting as a single routine.
@PROCESS DC(GAUS),OPT(3),VECTOR(LEVEL(2),REPORT(XLIST)), QPROCESS DIRECTIVE('*VD:')

```
c--.-- subroutine to compute the frechet derivatives for submission as
c--.- the response matrix to MEM_NL
c---- this subroutine uses finite difference perturbations
c----- uses Chuck Sword's XERR statistic
c-\cdots-- this routine is called for each ray involved in the data times ......
c--...
    subrout ine frechx(index, sloc1,tang1,sloc2,tang2,datatime, xerr,R)
c--.-- index = index number of ray as stored in geom file
c---- sloc1 = surface location of ray1
c-\cdots-- tang1 = take off angle of ray1
c.\cdots-- sloc2 = surface location of ray2
c-\cdots-- tang2 = take off angle of ray2
c--.-- datatime= actual measured time of ray
c--.-- xerr = Chuck Sword's statistic
c-\cdots...
c-\cdots-- n = number of rows in velocity grid
c-\cdots-- m = number of columns in velocity grid
c--.-- nint = spatial increment between grid rows
c--.-- mint = spatial increment between grid columns
c-----
c----- L = gaussian half-width (in metres)
c....- lc = constant multiplier of the model covariance matrix (use 1.0)
c--.-- bfx = (defunct variable - submit 1)
c----- bfz = (defunct variable - submit 1)
c-...-
c-\cdots.- nmax = maximum number of velocity nodes
c-\cdots--- mmax = maximum number of data rays
c-\cdots.- max = maximum of nmax and mmax
c
c----- -->common block /velocity/
c.-.-- vel(n,m)= values of velocity at grid points
c--..-
c----- -->common block /travel/
c-\cdots-- ray1travel( n*m,6) = array that stores cell,theta0, x,z,t for ray1's path
c---- num = number of points in ray1travel
c----- ray2travel(n*m,6) = array that stores cell,theta0, x, z,t for ray2's path
c\cdots--- num2 = number of points in ray2travel
c---.-
c---- i,j,k= temporary integer variables
c----- t1,t2,t3,t4,t5= temporary double-precision variables
c--...
c---- tri = coordinate indices of the apex of the current triangle
c----- row = the row of the current cell
c....- col = the column of the current cell
c----- trinode1,2,3 = velocity node indices of apices of current triangle
c---- rec1 = integer variable for record numbers
c\cdots--- R(ndat,n*m) = response matrix (frechet derivatives)
c-----
    real*4 nint,mint
    real*4 ray1travel,ray2travel
    real*4 sloc1,tang1,sloc2,tang2
    real*4 vel
    real*4 datatime,xerr,newxerr
    real*4 c,ggC
    real*4 lc
    real*4 R
    real*8 t1,t2, t3, t4,t5
    logical triflag
    logical affected
    integer*4 i,j,k
    integer*4 n,m
    integer*4 bfx,bfz
    integer*4 index
```

```
integer*4 L
integer*4 nmax,mmax,max
integer*4 num,num2
integer*4 row,col
integer*4 tri
integer*4 trinode1, trinode2,trinode3
integer*4 rec1
integer*4 p,mult,rem,block
parameter(n=41,m=41,nint=50.0e0,mint=50.0e0)
parameter(bfx=1,bfz=1)
parameter(nmax=1681, mmax=184, max=1681)
parameter(ndat=184)
dimension vel(n,m)
dimension ray1travel( }\mp@subsup{n}{}{*}m,6)\mathrm{ , ray2travel( (n*m,6)
dimension C(nmax, nmax),ggC(nmax,max)
dimension R(ndat, n*m)
dimension affected((n/bfz)*(m/bfx))
dimension tri(3,2)
common /velocity/vel
common /travel/ray1travel,num, ray2travel,num2
c
c---- set constants and initial variables
c
    t1=0.0d0
    t2=0.0d0
    t3=0.0d0
    t4=0.0d0
    t5=0.0d0
    do }1\textrm{i}=1,(n/bfz\mp@subsup{)}{}{*}(m/bfx
    affected(i)=.false.
1 continue
c---.- compute all velocity nodes affected by the rays
c----- begin working along path of ray1
c
    do 1000 i=1,num
c--.- first compute the 'tri' nodes affecting this cell
c
    p=int(ray1travel(i,1))
    rem=mod(p,(2*(m-1)))
    mult=(p-rem)/(2*(m-1))
    row=mult+1
    if(rem.eq.0) row=row-1
    if(rem.gt.0) then
        col=int((rem-1)/2.0)+1
    else
        col=m-1
        endif
    tri(1,1)=col
    tri(1,2)=row
    tri(2,1)=col+1
    tri(2,2)=row+1
    if(mod(ray1travel(i,1),2.0e0).eq.0.0e0) then
        tri (3,1)=col+1
            tri(3,2)=row
        else
            tri(3,1)=col
            tri(3,2)=row+1
            endif
c
c---- compute the corresponding velocity node indices
c
    trinode1=(tri(1,2)-1)*m+tri(1,1)
    trinode2=(tri(2,2)-1)*m+tri}(2,1
    trinode3=(tri(3,2)-1)*m+tri(3,1)
c
c-.-- compute the affected blocks
c
```

```
        do 500 j=1,3
        rem=mod(tri(j,1),bfx)
        mult=(tri(j,1)-rem)/bfx
        if(rem.eq.0) then
            col=mult
        else
            col=mul t+1
            endif
        rem=mod(tri(j,2),bfz)
        mult=(tri(j,2)-rem)/bfz
        if(rem.eq.0) then
            row=mul t
        else
            row=mul t+1
            endif
        k=(row-1)*m/bfx+col
        if(k.gt.n*m) k=n*m
        if(k.lt.1) k=1
        if(j.eq.1) trinode1=k
        if(j.eq.2) trinode2=k
        if(j.eq.3) trinode3=k
        continue
5 0 0
c
c----- set the required 'affected' flags
    affected(trinode1)=.true.
        affected(trinode2)=.true.
        affected(trinode3)=.true.
1000
continue
c
c---- continue computing the velocity nodes affected by the rays
c----- now do the same for ray2
c
    do 2000 i=1,num2
c
c--.-- first compute the 'tri' nodes affecting this cell
    p=int(ray2travel(i,1))
    rem=mod(p,(2*(m-1)))
    mult=(p-rem)/(2*(m-1))
    row=mult+1
    if(rem.eq.0) row=row-1
    if(rem.gt.0) then
        col=int((rem-1)/2.0)+1
        else
            col=m-1
            endif
        tri(1,1)=col
        tri(1,2)=row
        tri(2,1)=col+1
        tri(2,2)=row+1
        if(mod(ray2travel(i,1),2.0e0).eq.0.0e0) then
            tri(3,1)=col+1
            tri}(3,2)=ro
        else
            tri(3,1)=col
            tri(3,2)=row+1
            endif
c
c----- compute the corresponding velocity node indices
c trinodel=(tri(1,2)-1)*m+tri(1,1)
    trinode2=(tri(2,2)-1)*m+tri(2,1)
    trinode3=(tri(3,2)-1)*m+tri}(3,1
c
c-\cdots.-. compute the affected blocks
c
    do 1500 j=1,3
            rem=mod(tri(j,1),bfx)
            mult=(tri(j,1)-rem)/bfx
            if(rem.eq.0) then
```

```
                    col=mult
        else
            col=mult+1
            endif
        rem=mod(tri(j, 2),bfz)
        mult=(tri(j,2)-rem)/bfz
        if(rem.eq.0) then
            row=mul t
        else
            row=mul t+1
            endif
        k=(row-1)*m/bfx+col
        if(k.gt.n*m) k=n*m
        if(k.lt.1) k=1
        if(j.eq.1) trinode1=k
        if(j.eq.2) trinode2=k
        if(j.eq.3) trinode3=k
        continue
    1500
c
c----- set the required 'affected' flags
    affected(trinode1)=.true.
    affected(trinode2)=.true.
    affected(trinode3)=.true.
    continue
2000
c
c----- now begin raytracing through the perturbed velocity fields
    do 3000 i=1,(n/bfz)*(m/bfx)
    if(.not.affected(i)) goto 3000
c
c----- compute the top left node of the block
c
    rem=mod(i,m/bfx)
    mult=(i-rem)/(m/bfx)
        if(rem.eq.0) then
            col=m/bfx
            row=mult
        else
            col=rem
            row=mul t+1
            endif
            col=(col-1)*bfx+1
            row=(row-1)*bfz+1
c
c---- multiply affected velocities by }1.002
c
            do 3500 j=col,col+bfx-1
            do }3600\textrm{k}=\mathrm{ row, row+bfz-1
                vel(k,j)=vel(k,j)*1.0025e0
                                    continue
                    continue
3500
c
c-\cdots-- re-raytrace the ray
c
33 newxerr=0.0e0
    call xerrrt(sloc1,tang1, sloc2,tang2, datatime,newxerr)
c
c---- divide affected velocities by }1.0025\mathrm{ to restore them to original values
c
    do 3700 j=col,col+bfx-1
            do }3800\mathrm{ k=row, row+bfz-1
                vel(k,j)=vel(k,j)/1.0025e0
                t1=t1+dble(vel(k,j))
3800
                continue
            continue
            t 1=(t1/dble(bfx*bfz))*.0025d0
c
c---- put ((newxerr-xerr)/t1) into response matrix
c
    R(index,i)=((newxerr-xerr)/sngl(t1))
```


## c-... Loop back to do rest of the velocity nodes <br> 3000 continue <br> return <br> end

```
@PROCESS DC(GAUS),OPT(3),VECTOR(LEVEL(2),REPORT(XLIST)),
QPROCESS DIRECTIVE('*VD:')
c----- subroutine to compute the model covariance matrix for gaussian -----
c----- smoothing
c----- the half-width of the gaussian is 'L'
c----- the constant weighting of the covariance matrix is 'lc' .----
c----- subroutine also computes first/second order derivative components .-..-
    subroutine gauscalc(L)
c---- nq = n = number of rows in velocity grid
c----- mq = m = number of columns in velocity grid
c----- nint = spatial increment between grid rows
c----- mint = spatial increment between grid columns
c-----
c----- L = gaussian half-width (in metres)
c----- lc = constant multiplier of the model covariance matrix (use 1.0)
c--.--
c----- nmaxq = maximum number of velocity nodes
c---- mmaxq = maximum number of data rays
c---- maxq = maximum of nmaxq and mmaxq
c----C(0:nq-1,0:mq-1)=model covariance table defining the gaussian smoothing
c----- This is the output of this subroutine.
c-----
c----- i,j,k= temporary integer variables
c----- t1, t2, t3, t4, t5= temporary double-precision variables
c-----
    real*4 nint,mint
    real*4 nint2,mint2
    real*4 sum
    real*4 C
    real*4 lc
    real*8 t1, t2,t3, t4, t5
    integer*4 i,j,k
    integer*4 i1,i2,i3,i4,zz
    integer*4 L
    integer*4 n,nq,m,mq, ndat, dne
    integer*4 rem,mult
    integer*4 icol,irow, jcol,jrow
    integer*4 nmax,nmaxq,mmax,mmaxq,max,maxq
    parameter( }\textrm{nq}=90,\textrm{mq}=210
    parameter(nmaxq=17000, mmaxq =4200, maxq=17000)
    dimension C(0:nq-1,0:mq-1)
    common /conparm/n,m,nint,mint,nmax,mmax,max, ndat, dne
    common /gaus/C
c
c---- set constants and initial variables
    lc=1.0
    t1=0.0d0
    t2=0.0d0
    t3=0.0d0
    t4=0.0d0
    t5=0.00d0
    sum=0.0
    nint2=nint**2
    mint2=mint**2
c
c----- compute the model covariance table, C
c----- first compute the constant factor that is to be scaled relative to its
c----- distance away
c
    if((2.0*float(L)).le.mint) then
        t5=0.0d0
    else
```

```
        t5=dble(exp(-0.5e0*(mint/float(L))**2))
        endif
    c
    c---- if t5 is zero, set the C matrix equal to the identity
    c
        if(t5.eq.0.0d0) then
            do 90 i=0,n-1
            do }95\textrm{j}=0,\textrm{m}-
                    c(i,j)=0.0
                    continue
            continue
            C(0,0)=lc*1.0
            return
            endif
c
c-.--- now loop through for possible row and column differences
c
        do }100\textrm{i}=0,\textrm{n}-
            do }105\textrm{j}=0,\textrm{m}-
c
c----- calculate distance between points
                    t1=dble((float(i)*nint)**2)
                    t1=t1+dble((float(j)*mint)**2)
                    t1=dsqrt(t1)
c
c----- now compute relevant value of the gaussian
            if(t1.gt. 10.0*L) then
                    C(i,j)=0.0
                    goto }10
                    endif
                    t2=t1/dble(mint)
                    t2=t2**2
                    C(i,j)=lc*sngl(t5***2)
                    continue
                    continue
    return
    end
```

```
@PROCESS DC(GAUS),OPT(3),VECTOR(LEVEL(2),REPORT(XLIST)),
QPROCESS DIRECTIVE('*VD:')
c Brief Update History
c
MEMNL$5 - considerably modified jan 91, peter whiting
MEMNL$6 - considerably modified ->dec 91, peter whiting
MEMNL$8 - modified to read C matrix from dynamic common
                    ->dec 92, peter whiting
        - also convert to ESSL subroutines for the intensive
        processing -> jan 92, peter whiting
The notes directly below are Steve Brown's original comments.
    Maximum entropy image reconstruction subroutine.
This package implements the ME reconstruction algorithm presented in
J.Skilling and R.K.Bryan, (1984) Mon.Not.R.astr.Soc. 211, 111-124.
This routine solves for the maximum entropy image for both linear
and non-linear image reconstruction problems.
Each call to the routine MEM_NL takes one iteration from the current
image towards the maximum entropy image.
The program tries to follow the notation of Skilling and Bryan as
much as possible. All page and equation references in the program
comments refer to their paper.
Written: Steve Brown
    Department of Applied Mathematics
    University of Sydney
    NSW Australia.
Version 1. 13-October-89 (Not fully tested but works OK so far)
    subroutine MEMNL(n, m, I, D, F, dm, dn, A, chiO, Rchange,
    &
        entropy, chi2, test, nlin)
This routine is called by the driving program. It advances the
solution one iteration.
Input Parameters:
    n - number of image elements in the image vector I.
    m}\mathrm{ - number of data elements in the data vector D.
    I - array of n image values. On input I holds the current
            image. I corresponds to the f of Skilling and
            Bryan [Eqn 1].
            D - array of m experimental data values the image is to
                be fitted to [Eqn 2].
            F - for non-linear problems this is an array of m model
                data values corresponding to the image in I. For
                linear problems F is a work array and must be
                dimensioned to m}\mathrm{ or larger.
            R - response matrix. It has physical dimensions (dm,dn)
                with the first ( }m,n\mathrm{ ) containing the actual response
                matrix elements. The response matrix is defined as
                R(k,j)=dF(k)/dI(j)
            dm - dimensioned rows of array R.
            dn - dimensioned columns of array R.
            A - default intensity [Page 114, Eqn 6].
            chio - the desired chi-squared fit statistic [Eqn 4].
    Rchange - used only for linear image reconstruction. Rchange must
        be set to one on the first call and on any subsequent
        call where the response matrix, image vector or data
        vector have been changed. Rchange should be set to
        zero if these parameters are unchanged and the code will
```

c End of Steve Brown's comments.
c
c--.- nxx $=$ number of rows in discrete velocity model
c.-..- mxx = number of columns in discrete velocity model
c--.-. nmaxq = maximum number of velocity nodes
c----- mmaxq $=$ maximum number of data rays
c----- $\operatorname{maxq}=$ maximum of $n \max$ and $\operatorname{mmax}^{\max }$
c

c nmax is the largest allowed image size. mmax is the largest allowed c data size. dne sets the number of search directions used.

```
parameter(nmaxq=17000, mmaxq=4200,maxq=17000)
parameter (dneq = 4)
\begin{tabular}{|c|c|}
\hline real*4 & gS(nmaxq), gC(nmaxq), \(\mathrm{SO}, \mathrm{CO}, \mathrm{Itot}\), \\
\hline \& & sum1, sum2, Smu(dneq), Cmu(dneq), ganma(dn \\
\hline \& & e(nmaxq, dneq), x(dneq), almin, Caim \\
\hline
\end{tabular}
common /conparm/nxx,mxx,nint,mint, nmax, mmax, max, ndat, dne
```

c The gradients and values of S and C are stored in a common so their c values are preserved between subroutine calls.
common $/ \mathrm{mem} 5 \mathrm{~cm} / \mathrm{gS}, \mathrm{gC}, \mathrm{CO}, \mathrm{sO}$, Itot
if ( $(n$. gt. $n \max$ ) .or. ( $m$.gt. mmax)) then
print *, 'Error in MEM: nmax or max are too small'
stop
endif
c If first call or the response matrix $R$ has changed it is necessary to
c calculate the gradient information.
if ((Rchange .eq. 1) .or. ( $n$ lin .eq. 1)) then
call CALCGRAD ( $n, m, I, D, F, d m, d n, A, n l i n)$
call CALCTEST(n, test) endif
c Calculate the normalised search directions.

```
ne \(=\) dne
call SEARCH(e, ne, \(I, n, m, d m, d n)\)
```

c Now diagonalise the search subspace and construct quadratic models
c for $S$ and C.
call DIAGMOD(e, gamma, Smu, Cmu, ne, $n, m, d m, d n, ~ I)$
c Calculate the constants needed by the CONTROL subroutine.
call CONSTS(garma, ne, almin, Caim, chiO, CO, Cmu, Itot)
c Now calculate the increments x to move in each of the search
c directions.
call CONTROL( $\mathrm{x}, \mathrm{SO}, \mathrm{CO}, \mathrm{Smu}, \mathrm{Cmu}$, garma, Caim, almin, ne)
c Update the image vector along the search directions.
call UPDATE (x, e, ne, 1, n)
c Calculate the gradient information which is needed to calculate test.
if (nlin .ne. 1) then call CALCGRAD (n, m, I, D, F, dm, dn, A, nlin)
c Calculate test. call CALCTEST(n, test)
endif
entropy $=$ so
chi2 $=$ CO
return
end
@PROCESS DC(GAUS), OPT(3), VECTOR(LEVEL(2), REPORT(XLIST)), @PROCESS DIRECTIVE('*VD:')
c
$\qquad$
c
subroutine CALCGRAD(n, m, I, D, F, dm, dn, A, nlin)

integer $n \operatorname{maxq}, \operatorname{mmaxq}, \operatorname{maxq}$
integer $n \times x, m \times x, n d a t$, dne
integer*4 zz, fcb(20), rcb(20)
integer*4 i1, i2, i3, i4, vcount, Cind
logical set
real*4 nint, mint
real* $4 \quad A, I(n), D(m), F(m)$
real*4 t1, sumpl, tol
real*4 sig, ai,bi, XXerr, XXerr2, tvar
parameter ( $\mathrm{nq}=90, \mathrm{mq}=210$ )
parameter $\left(n_{\operatorname{maxq}}=17000\right.$, maxq $=4200$, maxq $=17000$ )
real* 4 Am(nmaxq), R(nmaxq)
real*4 gS(nmaxq), gC(nmaxq), S0, CO, sum1, sum2, sum3
\&
I tot, temp(nmaxq) , C , Ctmp
real*8 la,li
dimension C(0:nq-1,0:mq-1),Ctmp(nq*mq), Cind(nq*mq)
dimension tvar(maxq)
dimension sig(mmaxq), set(nmaxq)
dimension ai (maxq), bi (maxq), XXerr(mmaxq),XXerr2(maxq)
common $/ \mathrm{mem} 5 \mathrm{~cm} / \mathrm{gS}, \mathrm{gC}, \mathrm{CO}, \mathrm{SO}$, I tot
common /backg/Am
common /gaus/C
common /frechet/fcb, rcb
common/setvel/set
common/var/sig, ai, bi,XXerr, XXerr2
common /conparm/nxx, mxx, nint, mint, nmax, max, max, ndat, dne
tol $=1.0 \mathrm{e}-10$
c
c.-.- Finds the current chi-squared value
sum2 $=0.0$
do $30 \mathrm{k}=1$, m

$$
\text { sum2 }=\operatorname{sum} 2+((F(k)-D(k)) * * 2) * \operatorname{sig}(k)
$$

continue
$\mathrm{CO}=$ sum2
c Finds the current image entropy [Eqn 6] and gradient of $S$ [Eqn 7a].
I tot $=0.0$
sum1 $=0.0$
do $40 \mathrm{j}=1, \mathrm{n}$
$l a=d \log (d b l e(A m(j)))$
$\mathrm{li}=\mathrm{dlog}(\mathrm{dble}(\mathrm{I}(\mathrm{j})))$
$g S(j)=$ real $(l a-l i)$
sum1 $=$ sum1 $+I(j) *(\operatorname{gS}(j)+1.0)$
Itot $=I$ tot $+I(j)$
40 continue
S0 = sum1
c Finds the gradient of C [Eqn 8].
c.-.-- note the Covariance term (see TN. 10)
do $48 \mathrm{k}=1$, m
tvar $(k)=X \operatorname{Xerr}(k) * \operatorname{sig}(k) * 2.0+a i(k) * X X e r r 2(k)$
continue
do $49 \mathrm{j}=1, \mathrm{n}$
$\operatorname{temp}(j)=0.0$
continue
do $50 \mathrm{k}=1$, m
$\mathrm{x}=(\mathrm{k}-1) * \mathrm{n}$
call dmi014(R(1), n, x, rcb, fcb, 1, zz)
if(zz.ne.0) stop 'error reading frechet'
do $60 \mathrm{j}=1$, n
$\operatorname{temp}(j)=\operatorname{temp}(j)+R(j) * \operatorname{tvar}(k)$
continue
continue
if $(c(1,1)$.eq. 0.0$)$ then
cali scopy(n, temp, 1,gC,1)
return
endif
do $51 \mathrm{j}=1, \mathrm{n}$
if(set(j)) then
$g C(j)=0.0$
goto 51
endif
sum1 $=0.0$
sum3 $=0.0$
vcount $=0$
do $61 \mathrm{k}=1$, n
if(set(k)) goto 61
$t 1=C(\operatorname{int}((j-k) / m q), \operatorname{abs}(\bmod (j-1, m q)-\bmod (k-1, m q)))$
if(t1.eq.0.0) goto 61
vcount=vcount+1
Ctmp (vcount) $=\mathrm{t} 1$
Cind(vcount) $=k$
continue
sumf $=\operatorname{sdoti}$ (vcount $, \operatorname{Ctmp}(1), \operatorname{Cind}(1), \operatorname{temp}(1))$
sum3 $=\operatorname{sasum}$ (vcount $, \operatorname{Ctmp}(1), 1$ )
if(abs(sum1).gt.tol) then
$g C(j)=$ sum1/sum3
else
$g C(j)=0.0$

## endif

continue
return
end
@PROCESS DC(GAUS), OPT(3), VECTOR(LEVEL(2), REPORT(XLIST)), @PROCESS DIRECTIVE('*VD:')
c
c
subroutine $\operatorname{SEARCH}(e, n e, 1, n, m, d m, d n)$
integer ne, $n, m, n \max , \operatorname{mmax}, d n e, j, k, l, p, d m, d n$
integer $n \operatorname{maxq}$, maxq, maxq, dneq
integer ndat, $\mathrm{fcb}(20), \mathrm{rcb}(20)$
integer $x, y, z, \max , v, q, z z$
integer $\mathrm{i} 1, \mathrm{i} 2, \mathrm{i} 3, \mathrm{i} 4, \mathrm{i} 5, \mathrm{i} 6$, vcount, Cin .
integer* $4 \mathrm{nxx}, \mathrm{mxx}$
logical set
real*4 nint, mint, C, Ctmp
real*4 I( $n$ )
real*4 t1, t2, t3, t4, t5 , sum3
parameter ( $\mathrm{nq}=90, \mathrm{mq}=210$ )
parameter $\left(n_{\operatorname{maxq}}=17000\right.$, maxq $=4200$, $\operatorname{maxq}=17000$ )
parameter $($ dneq $=4)$
real*4 $g S(n \operatorname{maxq}), g C(n \operatorname{maxq}), S 0, C 0$, temp(nmaxq), temp2(mmaxq),
\&
temp3 (nmaxq), temp1 (nmaxq), I tot, e(nmaxq, dneq), sum1, sum2
real*4 sig(mmaxq), $R\left(n_{m a x q}\right)$
real*4 ai (maxq), bi (maxq),XXerr (mmaxq), XXerr2(maxq)
dimension $C(0: n q-1,0: m q-1), C t m p\left(n q^{*} m q\right), C i n d(n q * m q)$
dimension set (nmaxq)
common /mem $5 \mathrm{~cm} / \mathrm{gS}, \mathrm{gC}, \mathrm{CO}$, SO , I tot
common /var/sig, ai, bi, XXerr,XXerr2
common/setvel/set
common/gaus/C
common /frechet/fcb,rcb
common /conparm/nxx,mxx, nint, mint, nmax, max, max, ndat, dne
c Routine finds dne search directions according to Eqn 20 and the
c paragraph above Eqn 19.
c Finds gC The resulting search direction is normalised.
sum1 $=0.0$
call $\operatorname{scopy}(\mathrm{n}, \mathrm{gC}(1), 1, \mathrm{e}(1,1), 1)$
sum $1=\operatorname{snorm2}(\mathrm{n}, \mathrm{e}(1,1), 1)$
if (sum1 .eq. 0.0 ) then
sum1 $=1.0$
endif
do $20 \mathrm{j}=1, \mathrm{n}$
$e(j, 1)=e(j, 1) /$ sum1
continue
if (ne .eq. 1) return
c finds gS The resulting search direction is normalised.
sum1 $=0.0$
call $\operatorname{scopy}(\mathrm{n}, \mathrm{gS}(1), 1, \mathrm{e}(1,2), 1)$
sum1 $=$ snorm2 $(n, e(1,2), 1)$
if (sum1 .eq. 0.0 ) then
sum1 $=1.0$
endif
do $40 \mathrm{j}=1, \mathrm{n}$
$e(j, 2)=e(j, 2) /$ sum1
40
continue
c Loops around multiplying direction p-2 by matrix ggC to get
c direction p which is then normalised.
c Need to do this in four parts now to get ggC=C.Rt.R.C
do $50 p=3$, ne

```
if(C(1,1).eq.0.0) then
    call \(\operatorname{scopy}(n, e(1, p-2), 1\), temp1,1)
    goto 64
    endif
do \(60 \mathrm{k}=1\), n
    if(set(k)) then
            \(\operatorname{templ}(k)=0.0\)
            goto 60
            endif
    sum2 \(=0.0\)
    sum3 \(=0.0\)
    vcount \(=0\)
    do \(70 \mathrm{j}=1\), n
        if(set(j)) goto 70
            \(\mathrm{t} 1=\mathrm{C}(\operatorname{int}((j-k) / \mathrm{mq}), \operatorname{abs}(\bmod (j-1, \mathrm{mq})-\bmod (k-1, \mathrm{mq})))\)
            if(t1.eq.0.0) goto 70
            vcount=vcount+1
            Ctmp(vcount) \(=\mathrm{t} 1\)
            Cind(vcount) \(=\mathrm{j}\)
            continue
    sum2=sdot \(i(v c o u n t, \operatorname{Ctmp}(1), \operatorname{Cind}(1), e(1, p-2))\)
    sum3=sasum(vcount, \(\operatorname{Ctmp}(1), 1)\)
    temp1 \((k)=\) sum2/sum3
continue
do \(61 \mathrm{k}=1\), m
    sum2 \(=0.0\)
    \(\mathrm{x}=(\mathrm{k}-1) * \mathrm{n}\)
    call dmi014(R(1), n, x, rcb, fcb, 1, zz)
    if(zz.ne.0) stop 'error reading frechet'
    sum2 \(=\operatorname{sdot}\) ( \(n\), temp1, \(1, R, 1\) )
    temp2(k)=sum2*(2.0*sig(k)+4.0*ai(k)*XXerr(k)+
                bi(k)*XXerr2(k))
    continue
    do \(8872 \mathrm{j}=1\), n
    \(\operatorname{temp} 3(j)=0.0\)
    continue
    do \(62 \mathrm{k}=1\), m
        \(\mathrm{x}=(\mathrm{k}-1)\) *n
        call dmi014(R(1), n, x, rcb, fcb, 1, zz)
        if(zz.ne.0) stop 'error reading frechet'
        call syax ( \(n\), temp2(k),R(1), 1,temp3(1),1)
    cont inue
    if(c(1,1).eq.0.0) then
    call scopy ( n , temp3,1, temp, 1)
    goto 67
    endif
do \(63 \mathrm{k}=1\), n
    if(set(k)) then
                temp \((k)=0.0\)
            goto 63
            endif
        sum2 \(=0.0\)
        sum3 \(=0.0\)
        vcount \(=0\)
        do \(73 \mathrm{j}=1, \mathrm{n}\)
            if(set(j)) goto 73
            \(t 1=c(\operatorname{int}((j-k) / m q), \operatorname{abs}(\bmod (j-1, m q)-\bmod (k-1, m q)))\)
            if(t1.eq.0.0) goto 73
            vcount=vcount +1
            Ctmp(vcount) \(=t 1\)
            Cind(vcount) \(=\mathrm{j}\)
            continue
        sum2 \(=\) sdot \(i\) (vcount \(, \operatorname{Ctmp}(1), \operatorname{Cind}(1)\), temp3)
        sum3 \(=\) sasum (vcount, \(\operatorname{Ctmp}(1), 1\) )
        temp \((k)=\) sum2/sum3
    continue
```

sum1 $=0.0$

```
        call szaxpy(n,2.0,temp1(1),1,temp(1),1,e(1,p),1)
        sum1=snorm2(n,e(1,p),1)
        if (sum1 .eq. 0.0) then
        sum1 = 1.0
        endif
        do 100 j=1,n
        e(j,p) =e(j,p) / sum1
        continue
        continue
        return
        end
@PROCESS DC(GAUS),OPT(3),VECTOR(LEVEL(2),REPORT(XLIST)),
@PROCESS DIRECTIVE('*VD:')
c
c
subroutine DIAGMOD(e, gamma, Smu, Cmu, ne, n, m, dm, dn , I)
    integer ne, n, m, nmax, mmax, dne, j, k, ll, p, q, qq, tmp,
    &
        dm, dn, max
        integer i1,i2,i3,i4,zz,nmaxq,mmaxq,maxq, dneq,vcount,Cind
        integer nxx,mxx,ndat, x,fcb(20),rcb(20)
        logical set
        real*4 sumf,sumf2,twoA,sum3,L,LO
        real*4 nint,mint,C,Ctmp,t1
        parameter(nq=90,mq=210)
        parameter (nmaxq=17000,mmaxq=4200, maxq=17000)
        parameter (dneq = 4)
        real*4 I (nmaxq),I log(nmaxq),temp2(nmaxq, dneq)
        real*4 length(dneq)
        real*4 gS(nmaxq), gC(nmaxq), S0, CO, Itot, temp(mmaxq,dneq),
    & temp3(nmaxq,dneq),
    & sum1, sum2, g(dneq, dneq), MM(dneq, dneq), Smu(dneq),
& Cmu(dneq), gamma(dneq), e(nmaxq,dneq), V(dneq,dneq),
        eigval(dneq), e2(nmaxq,dneq)
    real*4 sig(mmaxq),R(nmaxq)
    real*4 ai(mmaxq),bi(mmaxq),XXerr(mmaxq),XXerr2(mmaxq)
    dimension C(0:nq-1,0:mq-1),Ctmp(nq*mq),Cind(nq*mq)
    dimension set(nmaxq)
    common /mem5cm/ gS, gC, CO, SO, Itot
    common /penalt/length,L,LO
    common /var/sig,ai,bi,XXerr,XXerr2
    common /setvel/set
    common /gaus/C
    common /frechet/fcb,rcb
    common /conparm/nxx,mxx, nint,mint,nmax,mmax,max, ndat, dne
    c Routine constructs and diagonalises the search subspace according to
    c the procedures in Eqn 24 and Section 3.7.1.
    c Calculate the metric tensor g [Eqn 24b].
1000 continue
        do }8\textrm{p}=1\mathrm{ , dne
            do }9\textrm{q}=1\mathrm{ , dne
                    g(p,q)=0.0
                    continue
            continue
        do }10p=1\mathrm{ , ne
            do 20 q = p, ne
                    sum1 = 0.0
                    do }30\textrm{j}=1,\textrm{n
                    sum1 = sum1 + e(j,p)*e(j,q)/I(j)
                    continue
                    g(p,q) = sum1
                    g(q,p) = sum1
                continue
        continue
```

c Diagonalise g. Eigenvectors are columns in the array V call JACOBI ( $g, n e$, dne, eigval, $V, q q$ )
c If any of the eigenvalues of $g$ are small then throw out that direction and repeat the calculation of $g$ etc. This protects against linear dependence in the search directions. NOTE the factor $1 . \mathrm{e}-4$ is an
c arbitrary choice.
$\mathrm{tmp}=0$
$p=1$
cont inue

1001
if (abs(eigval(p)).lt.amax1(eigval(1), eigval(2), eigval(3), $+\quad$ eigval(4))/10000.0) then
c + eigval (4), eigval(5), eigval(6))/10000.0) then do $40 \mathrm{q}=\mathrm{p}$, ne-1
call $\operatorname{scopy}(n, e(1, q+1), 1, e(1, q), 1)$
eigval $(q)=$ eigval $(q+1)$
continue
ne $=$ ne - 1
$\mathrm{tmp}=1$
if ( $p$.le. ne) goto 1001
endif
$p=p+1$
if ( $p$.le. ne) goto 1001
if (tmp .eq. 1) goto 1000
c Rescale the search directions so that the metric is cartesian. The
c new directions are held in the matrix e2.
do $60 \mathrm{j}=1, \mathrm{n}$
do $70 \mathrm{p}=1$, ne
sum1 $=0.0$
do $80 q=1$, ne
sum1 $=$ sum1 $+e(j, q) * V(q, p)$
80
continue
$e 2(j, p)=\operatorname{sum} 1 / \operatorname{sqrt}(\operatorname{abs}($ eigval(p)))
continue
continue
c Now calculate $M$ (held in matrix MM) [Eqn 24d].
c Need extra work to do R.C.e
do $18 \mathrm{p}=1$, dne
do $19 \mathrm{q}=1$, dne
$\operatorname{MM}(p, q)=0.0$
continue
continue
if(C( 1,1 ).eq. 0.0$)$ then
do $66 \mathrm{p}=1$, ne
call $\operatorname{scopy}(n, e 2(1, p), 1, \operatorname{temp} 3(1, p), 1)$
continue
goto 67
endif
do $90 \mathrm{k}=1$, n
if(set(k)) then
do $9191 \mathrm{p}=1$, ne
$\operatorname{temp} 3(k, p)=0.0$
cont inue
goto 90
endif
vcount=0
do $110 \mathrm{j}=1$, n
if(set(j).or.set(k)) goto 110
$t 1=C(\operatorname{int}((j-k) / m q), \operatorname{abs}(\bmod (j-1, m q)-\bmod (k-1, m q)))$
if(t1.eq.0.0) goto 110
vcount $=$ vcount +1
Ctmp(vcount) $=\mathrm{t} 1$
Cind(vcount) $=j$
continue
do $100 \mathrm{p}=1$, ne
sumf $=\operatorname{sdot} i(v$ count $, \operatorname{Ctmp}(1), \operatorname{Cind}(1), e 2(1, p))$
sum3 $=$ sasum(vcount $, \operatorname{Ctmp}(1), 1$ )
$\operatorname{temp} 3(k, p)=$ sum1/sum3

```
100 continue
9 0
    continue
    do 91 k = 1,m
            x=(k-1)*n
            call dmi014(R(1),n,x,rcb,fcb,1,zz)
            if(zz.ne.0) stop 'error reading frechet'
            do 101 p=1, ne
                sum1=sdot (n,R(1),1,temp3(1,p),1)
                    temp(k,p)=sum1*(2.0*sig(k)+4.0*ai(k)*XXerr(k)+
                    bi(k)*XXerr2(k))
l01
            continue
    continue
    do }120\textrm{p}=1\mathrm{ , ne
            do }130q=p\mathrm{ , ne
                sum1=sdot (m,temp(1,p),1,t\operatorname{temp}(1,q),1)
                    sum2=sdot (n,e2(1,p),1,temp3(1,q),1)
                    MM (p,q) = sum1+2.0*sum2
                    MM(q,p) = MM(p,q)
            continue
    continue
        write(6,'(2x,a14,i5)') 'ne is now (c) ',ne
c Diagonalise M. Eigenvectors are columns in the array V
    call JACOBI (MM, ne, dne,gamma,V,qq)
c Rescale the search directions so that M is diagonal. The
c new directions are held in the matrix e.
    do }150\textrm{j}=1\mathrm{ , n
        do }160p=1\mathrm{ , ne
            sum1 = 0.0
            do }170\textrm{q}=1\mathrm{ 1, ne
                        sum1 = sum1 + e2(j,q)*V(q,p)
1 7 0
                    continue
                        e(j,p) = sum1
            continue
160
    continue
```

c
c---- compute lengths of these final search directions to be passed
c---- to CONTROL for correct distance penalty implementation
c
do $155 \mathrm{p}=1$, ne
length $(p)=\operatorname{snorm} 2(n, e(1, p), 1)$
155
continue
c Calculate the final components of the quadratic models Smu and Cmu
c [Eqns 24a and 24c].
do $180 p=1$, ne
$\operatorname{smu}(p)=\operatorname{sdot}(n, e(1, p), 1, g s(1), 1)$
$\operatorname{Cmu}(p)=\operatorname{sdot}(n, e(1, p), 1, g C(1), 1)$
cont inue
return
end
@PROCESS DC(GAUS), OPT(3), VECTOR(LEVEL(2), REPORT (XLIST)), @PROCESS DIRECTIVE('*VD:')
c
c
subroutine CONSTS(gamma, ne, almin, Caim, chio, CO, Cmu, Itot)

```
    integer ne, p, dne, dneq
    integer nxx,mxx,nmax,mmax,max,ndat
    real*4 chi0,chiOold
    real*4 nint,mint
    real*4 gamma(ne), Cmu(ne), almin, Caim, LO, CO, Itot,
    & sum1, Cmin,L
    real*4 dlimit
```

    parameter (dneq=4)
    real*4 length(dneq)
common /dist/dlimit
cormon /penalt/length,L,LO
common /conparm/nxx,mxx, nint,mint, nmax, mmax, max, ndat, dne
c Routine calculates the constants needed by the control procedure.
c Find Cmin [Eqn 28].
sum1 $=0.0$
do $10 p=1$, ne
sum1 $=$ sum1 $+(\mathrm{Cmu}(p) * * 2) /$ gamma $(p)$
continue
$\mathrm{Cmin}=\mathrm{CO}-0.5 *$ sum1
c Find Caim [Eqn 29].
c Skilling and Bryans choice of 0.667 and 0.333 .
Caim $=0.667 * \mathrm{Cmin}+0.333 * \mathrm{CO}$
if (chio .gt. Caim) Caim $=$ chio
if(Caim.gt.CO) Caim=CO
c Find LO [Paragraph above Eqn 28].
c *** I changed this from using Itot to using nmax, which ***
*** I feel is a more useful limit. Peter W 13/03/91 ***
$\mathrm{LO}=(2.0 * \mathrm{nmax}) / \mathrm{dl}$ imit
c Find alpha min [Paragraph below Eqn 32].
almin $=$ - gamma(1)
do $20 \mathrm{p}=2$, ne
sum1 $=-\operatorname{gamma}(p)$
if (sum1 .gt. almin) almin = sum1
continue
if (almin .lt. 0.0 ) almin $=0.0$
return
end
@PROCESS DC(GAUS), OPT(3), VECTOR(LEVEL(2), REPORT(XLIST)),
@PROCESS DIRECTIVE('*VD:')
c
$\qquad$
subroutine CONTROL( $\mathrm{x}, \mathrm{sO}, \mathrm{CO}, \mathrm{Smu}, \mathrm{Cmu}$, gamma, Caim, almin, ne)
integer ne, $q$, asuc, psuc, afinished, pfinished, dne, dneq
integer $n x x, m \times x, n m a x, m m a x, m a x, n d a t$
parameter $($ dneq $=4)$
real*4 length(dneq)
real*4 nint,mint
real*4 $x(n e)$, $50, C 0, S m u(n e), C m u(n e)$, gamma(ne), Caim, L0,
\& almin, alpha, p, C, Cp, L, xs(dneq), alow, plow, ahigh,
\& phigh, Caimh
real*4 Lsuc,Csuc
common /penal t /length, L,LO
common /conparm/nxx,mxx, nint,mint, nmax,mmax,max,ndat, dne
c Implements the control procedure charted on Page 122, Figure 3 and
c described in Sections 3.7.2 and 3.7.3.
if (CO .lt. Caim*1.01) then Caimh $=$ Caim*1. 01
else
Caimh $=$ Caim
endif
$p=0.0$
plow $=0.0$
phigh $=-1.0$
psuc $=0$
pfinished $=0$
c Start of the P chop. 10 continue

```
    alow = almin
    ahigh = -1.0
    alpha = almin + 1.0
    asuc =0
    afinished = 0
    c Start of the alpha chop
    20 continue
    c Calculate C [Eqn 23b], Cp [Eqn 35], L [Eqn 27c] and x [Eqn 34].
    C = CO
    Cp}=\textrm{CO
    L}=0.
    do 1 q = 1, ne
        x(q) = (alpha*Smu(q) - Cmu(q)) / (p+gamma(q)+alpha)
        c = c + Cmu(q)*x(q) + 0.5*gamma(q)*(x(q)**2)
        Cp = Cp + Cmu(q)*x(q) + 0.5*(p+gamma(q))*(x(q)**2)
        L = L + length(q)*x(q)**2
    continue
    if ((CP .gt. CO) .and. (Cp .gt. Caimh)) then
            ahigh = alpha
        alpha = 0.5*(alpha + alow)
    elseif(L.gt.LO) then
        if(Cp.le.CO) then
            alow=alpha
            if(ahigh.gt.0.0) then
                atpha=0.5*(ahigh+alpha)
            else
                    alpha=2.0*alpha
                    endif
        else
            ahigh=alpha
            alpha=0.5*(alpha+alow)
            endif
        elseif(c.lt.Caim) then
        if(Caimh.ge.CO) then
            do 2 q = 1, ne
                    xs(q) = x(q)
                    continue
                    asuc=1
                    Lsuc=L
                    Csuc=C
                    endif
        alow=alpha
        if(ahigh.gt.0.0) then
            alpha=0.5*(ahigh+alpha)
        else
            alpha=2.0*alpha
            endif
        else
        do 4 q=1, ne
            xs(q) = x(q)
            continue
        asuc=1
        Lsuc=L
        Csuc=C
        ahigh=alpha
        alpha=0.5*(alow+alpha)
        endif
    if ((ahigh.gt.0.0) .and. ((ahigh-alow).lt.
            (1.0E-3*ahigh+1.0E-10))) afinished = 1
        if (alpha .gt. 1.0E20) afinished = 1
        if (afinished .ne. 1) goto 20
c End of alpha chop
    if (asuc .eq. 1) then
    psuc = 1
    phigh = p
```

```
            p=0.5*(plow + p)
            else
            plow = p
            if (phigh .gt. 0.0) then
                    p=0.5*(phigh + p)
            else
                    p=2.0*(p + 1.0)
            endif
        endif
                            if ((psuc.eq.1) .and. ((p.eq.0.0) .or. (phigh-plow).lt.
    &
                (1.0E-3*phigh+1.0E-10))) pfinished = 1
c check for occasional bad topology (rare)
    if((psuc.eq.0).and.(p.gt.1.0e20)) then
            pfinished=1
            write(2,'(2x,a27)') 'P chop blow-up encountered'
            endif
        if (pfinished .ne. 1) goto }1
    c End of P chop
    c Topology correction
        if(psuc.eq.0) then
            do }5\textrm{q}=1,\textrm{ne
                x(q)=1.0e-2*L0*(Smu(q) -Cmu(q))
                    continue
            endif
c Copy the most recent successful x (array xs) to the output increment
c array x.
3333 do 3q=1, ne
            x(q) = xs(q)
        continue
    WRITE(2,'(2X,A36,G16.8)')'MEM_NL$: expected Chi in CONTROL - ',Cp
    WRITE(2,'(2X,A36,G16.8)')'MEM_NL$: expected ENT in CONTROL - ',S
    WRITE(2,'(2X,A36,G16.8)')'MEM_NL$: calculated distance - ',L
    WRITE(2,'(2X,A36,G16.8)')'MEM_NL$: maximum distance - ',LO
    WRITE(2,'(2X,A26,G16.8)')'MEM_NL$: minimum alpha - ', almin
    WRITE(2,'(2X,A26,G16.8)')'MEM_NL$: final alpha - ',alpha
    return
    end
@PROCESS DC(GAUS),OPT (3),VECTOR(LEVEL(2),REPORT(XLIST)),
@PROCESS DIRECTIVE('*VD:')
c
c-----------------------------------------------------------------------------------
c
subroutine UPDATE(x, e, ne, I, n)
```



```
c Update the image vector [Eqn 25].
    sum=0.0
    do 10 p = 1, ne
        do 20 j=1,n
            I(j) = I(j) + x(p)*e(j,p)
            sum=sum+(x(p)*e(j,p))**2
            continue
    continue
    write(6,'(2x,a25,g16.8)') 'actual distance of update',sqrt(sum)
```

Protect against stray negative image values. do $30 \mathrm{j}=1$, n if $(I(j)$.le. 0.0$) I(j)=1 . e-6$
30 continue
return end
@PROCESS DC(GAUS), OPT(3), VECTOR(LEVEL(2), REPORT(XLIST)),
@PROCESS DIRECTIVE('*VD:')
c
c
subroutine CALCTEST ( $n$, test )

```
integer n, nmax, j,mmax
integer nmaxq,mmaxq,maxq
integer nxx,mxx,max,ndat,dne
real*4 test
real*4 nint,mint
parameter(nmaxq=17000,mmaxq=4200,maxq=17000)
real*4 gS(nmaxq), gC(nmaxq), S0, C0, I tot, sum1, sum2, sum3
common /mem5cm/ gS, gC, CO, SO, I tot
common /conparm/nxx,mxx, nint,mint,nmax,mmax,max,ndat, dne
```

c Finds skilling and Bryans test parameter [Eqn 37$].$
sum1 $=0.0$
sum2 $=0.0$
do $10 \mathrm{j}=1, \mathrm{n}$
sum1 $=$ sum1 $+\operatorname{gS}(j) * * 2$
sum2 $=$ sum2 $+g C(j) * * 2$
10 continue
sum1 $=$ sqrt(sum1)
sum2 $=\operatorname{sqrt}($ sum2 $)$
sum3 $=0.0$
if ((sum1 .ne. 0.0 ) , and. (sum2 .ne. 0.0$)$ ) then
do $20 \mathrm{j}=1, \mathrm{n}$
sum $3=\operatorname{sum} 3+(g s(j) / \text { sum } 1-g C(j) / \text { sum2 })^{* * 2}$
continue
endif
test $=0.5 *$ sum3
return
end
@PROCESS DC(GAUS), OPT (3), VECTOR(LEVEL(2), REPORT (XLIST)), @PROCESS DIRECTIVE('*VD:')
c
c
SUBROUTINE JACOBI (A,N,NP,D,V,NROT)
implicit real*4 (a-h,o-z)
implicit integer (i-n)
PARAMETER (NMAX=4)
DIMENSION A(NP,NP), D(NP),V(NP,NP),B(NMAX),Z(NMAX)
DO $12 I P=1, N$
DO $11 \mathrm{IQ}=1, \mathrm{~N}$
$V(I P, I Q)=0$.
CONTINUE
$V(I P, I P)=1$.
CONTINUE
DO 13 IP $=1$, $N$
$B(I P)=A(I P, I P)$
$D(I P)=B(I P)$
$Z(I P)=0$.
CONTINUE
NROT $=0$
DO $24 \mathrm{I}=1,50$
SM=0.
DO 15 IP $=1, \mathrm{~N}-1$
DO $14 I Q=I P+1, N$
$S M=S M+A B S(A(I P, I Q))$

CONTINUE
CONTINUE
IF (SM.EQ.O.)RETURN
IF(I.LT.4)THEN
TRESH $=0.2 * S M / N^{* *} 2$
ELSE
TRESH=0.
ENDIF
DO $22 I P=1, N-1$
DO $21 I Q=I P+1, N$
$\mathrm{G}=100$. *ABS(A(IP,IQ))
IF((I.GT.4).AND.(ABS(D(IP))+G.EQ.ABS(D(IP)))
.AND. (ABS $(D(I Q))+G . E Q . A B S(D(I Q))))$ THEN
$A(I P, I Q)=0$.
ELSE IF(ABS (A(IP,IQ)).GT.TRESH)THEN
$H=D(I Q)-D(I P)$
IF (ABS (H) +G.EQ.ABS (H) )THEN
$T=A(I P, I Q) / H$
ELSE
THETA $=0.5 * H / A(I P, I Q)$
$T=1 . /($ ABS (THETA)+SQRT(1.+THETA**2))
IF(THETA.LT.O.) $T=-T$
ENDIF
$\mathrm{C}=1 . / \operatorname{SQRT}\left(1+\mathrm{T}^{* *} 2\right)$
$\mathrm{S}=\mathrm{T}$ * C
$T A U=S /(1 .+C)$
$H=T * A(I P, I Q)$
$Z(I P)=Z(I P)-H$
$Z(I Q)=Z(I Q)+H$
$D(I P)=D(I P)-H$
$D(I Q)=D(I Q)+H$
$A(I P, I Q)=0$.
DO $16 \mathrm{~J}=1, \mathrm{IP}-1$
$G=A(J, I P)$
$H=A(J, I Q)$
$A(J, I P)=G-S^{*}(H+G * T A U)$
$A(J, I Q)=H+S^{*}\left(G-H^{*} T A U\right)$
CONTINUE
DO $17 \mathrm{~J}=\mathrm{IP}+1, \mathrm{IQ}-1$
$\mathrm{G}=\mathrm{A}(\mathrm{IP}, \mathrm{J})$
$H=A(J, I Q)$
A (IP, J) $=\mathrm{G}-\mathrm{S}^{*}\left(\mathrm{H}+\mathrm{G}^{*} \mathrm{TAU}\right)$
$A(J, I Q)=H+S^{*}\left(G-H^{*} T A U\right)$
CONTINUE
DO $18 \mathrm{~J}=\mathrm{IQ}+1, \mathrm{~N}$
$\mathrm{G}=\mathrm{A}(\mathrm{IP}, \mathrm{J})$
$H=A(I Q, J)$
$A(I P, J)=G-S^{*}(H+G * T A U)$
$A(I Q, J)=H+S^{*}\left(G-H^{*} T A U\right)$
CONTINUE
DO $19 \mathrm{~J}=1, \mathrm{~N}$
$G=V(J, I P)$
$H=V(J, I Q)$
$V(J, I P)=G-S^{*}(H+G * T A U)$
$V(J, I Q)=H+S^{*}\left(G-H^{*} T A U\right)$
CONTINUE
NROT $=$ NROT +1
ENDIF
CONTINUE
CONTINUE
DO $23 \mathrm{IP}=1, \mathrm{~N}$
$B(I P)=B(I P)+Z(I P)$
$D(I P)=B(I P)$
$Z(I P)=0$.
CONTINUE
CONTINUE
PAUSE '50 iterations should never happen'
RETURN
END

```
@PROCESS DC(GAUS),OPT(3),VECTOR(LEVEL(2),REPORT(XLIST)),
```

QPROCESS DIRECTIVE('*VD:')
c----- program to perform maximum entropy reflection tomographic imaging
c--... utilises the 'directional reception' method of ray tracing
c----- uses Chuck Sword's XERR statistic
c----- uses Steve Brown's MEM NL subroutine for the ME optimisation ......
c----- updates to MEM_NL for smoothing
c----- all real calculations in double-precision
c--.-- modified to use a tipex scratch file for the frechet matrix -...-
c program metomos
c--..- ndatq $=$ maximum number of data rays used in inversion
c----- ndat $=$ actual number of data rays
c-----
c.-... nq $=$ maximum number of rows in velocity grid
c----- mq $=$ maximum number of columns in velocity grid
c--..- $n=$ actual number of rows in velocity grid
c.-..- m actual number of columns in velocity grid
c-----
c--.-- nmaxq $=$ maximum number of velocity nodes
c----- mmaxq $=$ maximum number of data rays
c---- $\operatorname{maxq}=$ maximum of $n \operatorname{maxq}$ and maxq
c----- nmax $=$ actual number of velocity nodes
c----- mmax $=$ actual number of data rays
c----- $\max =$ maximum of $n \max$ and max
c-----
c----- nint $=$ spatial increment between grid rows
c----- mint $=$ spatial increment between grid columns
c----
c---- chio = desired chi value
c----- LL $\quad$ gaussian half-width (in metres)
c---- vel(nq,mq)= values of velocity at grid points
c----- $A m(n \operatorname{maxq})=$ the image vector storing the prior information
c----- $\operatorname{Rray}\left(n q^{*}\right.$ mq) $=$ temporary vector to hold Frechet derivative values
c----- sloc1 $=$ source location of first ray
c---- tang $1=$ take-off angle in radians (positive angle implies
c----- in direction of increasing distance) of first ray
c----- sloc2 $=$ source location of second ray
c---- tang2 $=$ take-off angle in radians (positive angle implies
c----- in direction of increasing distance) of second ray
c----- xerr $=$ Chuck Sword's statistic
c---- i, j,k= temporary integer variables
c---- $\mathrm{t} 1, \mathrm{t} 2, \mathrm{t} 3, \mathrm{t} 4, \mathrm{t} 5=$ temporary double-precision variables
-.....
c--.-- chimem(5) $=$ vector for remembering latest iterations for
c---- automatic stage changes, chi0 determination, and
c----- termination
c--.-- Image(nq*mq+5)= vector of velocity values
c----- D(ndatq) = data travel times
c----- zero(ndatq) = array storing zero
c----- $F$ (ndatq) $\quad$ current model travel times
c.--- chi2 $=$ chi value at start of iteration
c..... entropy $=$ value of entropy at start of iteration
c---- test $\quad=$ Skilling and Bryan's test statistic at start of iteration
c--...
c.-..- raybad $=$ logical vector to remember the rays that had Xerr's
c--..- greater than the threshold.
c--.- sig(ndatq) $=$ holds the ray weights, computed by VARIAN
c---- ai(ndatq), bi(ndatq) $=$ assist in the application of the weights
c-....
c--.-- numnull $=$ records the number of nulled rays ie xerr>threshold in XERRRT
c--.-- numturn $=$ records the number of turned rays nulled in XERRRT
c-----

```
set(n*m) = logical array set to .true. is vel node not to change
c----- ucount = 'unstability count'. incremented when dlimit increased
c-----
real*4 sloc1,tang1,xerr
real*4 sloc2,tang2
real*4 nint,mint
real*4 ray1travel,ray2travel
real*4 Rray,D,zero, F, Fold
real*4 Image,Am
real*4 chi0,chi2,entropy, test,testold
real*4 dlimit,pctd
real*4 length,L,LO
real*4 vel
real*4 lc
real*4 sig,ai,bi,XXerr,XXerr2
real*4 t11, t12,t13,t14,t15
real*8 t1, t2, t3, t4, t5
real*8 tol,pi,pion2, twopi
logical raybad
logical set
integer*4 set1,set2,set3,set4,dskip,stop,maxit,form
integer*4 stopc
integer*4 i, j, k, x, dne, dneq
integer*4 n,nq,m,mq,nbym
integer*4 num,num2
integer*4 numnull, numturn, ucount,iter,numtot
integer*4 ndat,ndatq
integer*4 LL,LLn,LLm
integer*4 nmax,nmaxq,mmax,mmaxq,max,maxq
integer*4 total
integer*4 rem,mult,row,col,block,nc
integer*4 tImage
integer*4 acnm,fcb,rcb
parameter(ndatq=4200)
parameter ( }\textrm{nq}=90,\textrm{mq}=210\mathrm{ ) 
parameter(nmaxq=17000,mmaxq=4200,maxq=17000)
parameter(dneq=4)
real*4 gS(nmaxq),gC(nmaxq),SO,CO,I tot
dimension vel(nq,mq)
dimension chimem(5)
dimension Image(nq*mq+5)
dimension raybad(ndatq)
dimension set(nq*mq)
dimension length(dneq)
dimension Am(nmaxq)
dimension tImage(nq*mq+5)
dimension ray1travel( nq*mq,6),ray2travel( nq*mq,6)
dimension Rray(nq*mq)
dimension D(ndatq), zero(ndatq), F(ndatq), Fold(ndatq)
dimension sig(ndatq)
dimension ai(ndatq),bi(ndatq),XXerr(ndatq),XXerr2(ndatq)
dimension sloc1(ndatq),sloc2(ndatq),tang1(ndatq),tang2(ndatq)
dimension C(0:nq-1,0:mq-1)
dimension acnm(2),fcb(20),rcb(20)
common /velocity/vel
common /iterat/iter
common /gausiz/LL,LLn,LLm
common /backg/Am
cormmon/constant/tol,pi,pion2, twopi
common /dist/dlimit
common /lgxerr/raybad
common /var/sig,ai,bi,XXerr,XXerr2
common /nums/numnull,numturn
common /setvel/set
common /gaus/C
common /frechet/fcb,rcb
common /penalt/length,L,LO
```

```
    common /conparm/n,m,nint,mint,nmax,mmax,max,ndat,dne
    cormon /mem5cm/gS,gC,CO,SO,I tot
    common /temp/nc
    data acnm/'SCRA','TCH '/
c
c----- read variables from FT21F001
c---- set1-set2 and set3-set4 and 'set' velocity nodes (inclusive)
c----- dskip is the number of data cards to skip before each read
c---- submit -1 for LL and 0 for sets and dskip to turn off
c----- submit 1 for 'stop' to stop inversion after one iteration
c---.- 'maxit' is the max iterations in a single stage
c----- 'potd' defines the percentage drop for chi control
c----- 'form' =0 gives 5g12.6 format for data, =1 gives 5g15.6
c
    read(21,1021) n,m,nint,mint,nmax,mmax,max, ndat,dne
    read(21, 1022) LL,set1,set2,set3,set4,dskip,stop,maxit,pctd,form
    write(6,1023) 'n = ',n
    write(6,1023) 'm= 'm
    write(6,1024) 'nint = ', nint
    write(6,1024)'mint = ',mint
    write(6,1023) 'nmax = ',nmax
    write(6,1023) 'mmax = ',mmax
    write(6,1023) 'max = ',max
    write(6,1023) 'ndat = ',ndat
    write(6,1023) 'dne = ',}\mathrm{ ,dne
    write(6,1023)'LL = ',LL
    write(6,1023) 'set1 = ', set1
    write(6,1023) 'set2 = ',}\mathrm{ ,set2
    write(6,1023) 'set3 = ', set3
    write(6,1023) 'set4 = ', set4
    write(6,1023) 'dskip= ',dskip
    write(6,1023) 'stop = ',stop
    write(6,1023) 'maxit= ',maxit
    write(6,1024) 'pctd = ',pctd
    write(6,1023)'form = ',}\mathrm{ ,form
    if(n.gt.nq) stop 'n .gt. default maximum'
    if(m.gt.mq) stop 'm .gt. default maximum'
    if(nmax.gt.nmaxq) stop 'nmax .gt. default maximum'
    if(mmax.gt.mmaxq) stop 'mmax .gt. default maximum'
    if(max.gt.maxq) stop 'max .gt. default maximum'
    if(ndat.gt.ndatq) stop 'ndat .gt. default maximum'
    if(dne.gt.dneq) stop 'dne .gt. default maximum'
    if(LL.eq.-1) LL=int(mint)*(m-1)
1021 format(2i5,2f10.2,5i5)
1022 format(8i5,f10.2,i5)
1023 format(2x, a7, i5)
1024 format(2x,a7,f12.2)
c
c--.-- set constants and initial variables
numtot=0
    nc=0
    stopc=0
    LLn=10* int(LL/nint)+1
    LLm=10*int(LL/mint)+1
    numnul l=0
    numturn=0
    iter=0
    ucount=0
    dlimit=1.0
    lc=1.0
    chi0=0.0
    nbym=n*m
    tol=1.0d-10
    pi=4.0d0*datan(1.0d0)
    pion2=2.0d0*datan(1.0d0)
    twopi=2.0d0*pi
    t1=0.0d0
    t2=0.0d0
    t3=0.0d0
    t4=0.0d0
```

```
    t5=0.0d0
    chi2=0.0e0
    entropy=0.0e0
    test=0.0e0
    testold=0.0e0
    total=0
    do }6\mathrm{ i=1,ndat
        raybad(i)=.false.
        sig(i)=1.0
        F(i)=9999.0
            continue
    do }7\textrm{i}=1\mathrm{ ,nbym
    if(((i.ge.set1).and.(i.le.set2)).or.
            ((i.ge.set3).and.(i.le.set4))) then
            set(i)=.true.
        else
            set(i)=.false.
            endif
    continue
c
c----- read the velocity cards
    do }610\textrm{i}=1,(\operatorname{int}(\mp@subsup{n}{}{*}m/10)-1)*10+1,1
        read(10,'(10i6)') (tImage(k), k=i,i+9)
        write(6,'(10i6)') (tImage(k), k=i,i+9)
            continue
    read(10,'(10i6)') (tImage(j), j=int(n*m/10)*10+1, n*m,1)
    do }611\textrm{i}=1\mathrm{ ,nbym
        Image(i)=float(tImage(i))
            continue
        do 612 i=1, (int(n*m/10)-1)*10+1,10
            read(12,'(10(1x,i5))') (tImage(k),k=i,i+9)
            write(6,'(10(1x,i5))') (tImage(k),k=i,i+9)
            continue
        read(12,'(10i6)') (tImage(j),j=int(n*m/10)*10+1, n*m, 1)
    do }613\textrm{i}=1\mathrm{ ,nbym
        Am(i)=float(tImage(i))
            continue
        do }100\textrm{i}=1,\textrm{n
            do }110\textrm{j}=1,\textrm{m
                vel(i,j)=Image((i-1)*m+j)
                continue
            continue
100
    open the Frechet derivative scratch file
c
    call opn013(fcb, 1, acnm,3,0,0,0,4096,i)
    if(i.ne.0) stop 'error opening scratch'
c
c----- open the log files
c
    open(unit=2,file='metlog')
    open(unit=5, file='metfil')
c
c----- read the data times into an array
c----- this file contains the two initial angles, the two surface locations
c----- and the data time
c
    do }50\textrm{i}=1\mathrm{ ,ndat
        if(dskip.gt.0) then
            do }51\textrm{j}=1\mathrm{ ,dskip
                if(form.eq.0) then
                    read(9,'(g12.6)') t15
                else
                    read(9,'(g15.6)') t15
                        endif
                continue
            endif
        if(form.eq.0) then
        read(9,'(5g12.6)') sloc1(i),sloc2(i),tang1(i),tang2(i),D(i)
        write(6,'(5g12.6)') sloc1(i),sloc2(i),tang1(i),tang2(i),D(i)
```

```
        else
        read(9,'(5g15.6)') sloc1(i),sloc2(i),tang1(i),tang2(i),D(i)
        write(6,'(5g15.6)') sloc1(i),sloc2(i),tang1(i),tang2(i),D(i)
        endif
        zero(i)=0.0e0
        continue
    rewind(9)
    c
    c-.--- calculate the gaussian smoothing through the model covariance matrix
    c----- STAGE LOOPING POINT
    10101 call gauscalc(LL)
        write(2,'(2x,a25,i6)') 'current half-width (LL) = ',LL
        do }55\textrm{i}=1,
        chimem(i)=0.0
        continue
5 5
c
c----- ItERATION LOOPING POINT
1 continue
    numnul l=0
    numturn=0
    do }3\textrm{i}=1,\mathrm{ ndat
        Fold(i)=F(i)
3 continue
c
c
    do }1000\textrm{i}=1\mathrm{ , ndat
    if(raybad(i)) then
                goto 1000
                endif
            xerr=0.0e0
c
c----- zero the frechet derivative temporary vector Rray
c---- this vector must be zero when each ray is traced
c
    do 61 j=1,nbym
        Rray(j)=0.0e0
61 continue
c
c----- call raytracing subroutine
c
    call xerrrt(sloc1(i),tang1(i),sloc2(i),tang2(i),D(i),xerr)
c
c----- call subroutine to calculate frechet derivatives
c
            if(xerr.ne.0.0) then
                call sade2(i, sloc1(i),tang1(i),sloc2(i),tang2(i),D(i),
                endif
c
c----- update model xerr in F array
c
    F(i)=xerr
c--.- zero the frechet derivatives of any set nodes for this ray
c
    do 305 j=1,nbym
            if(set(j)) Rray(j)=0.0
            continue
305
c
c----- write Rray temporary vector out to direct access disc file
c
    j=(i-1)*nbym
    call dmo015(Rray(1),nbym, j,rcb,fcb,1,k)
    if(k.ne.0) stop 'error writing frechet'
c
c----- loop back to do rest of rays
1000
    continue
```

```
    write(2,'(2x,a14,i10)')''nulled rays = ',numnull
    write(2,'(2x,a21,i10)') 'nulled turned rays = ',numturn
    write(2,'(2x,a21,i10)') ''number count SADE2 = ',nc
    nc=0
c
c---- compute the weights based on the rays Xerr's
c
    call varian(F)
c---- call the ME optimisation subroutine
c
    call MEMNL(nbym,ndat, Image,zero,F,ndat,
        +
c
c---- convert the the new Image vector into the vel matrix for output
c
    do }320\quadi=1,
        do }330\textrm{j}=1,\textrm{m
            vel(i,j)=Image((i-1)*m+j)
            continue
330 contin
320
c----- write results out to log file
    write(2,'(2x,a16,g16.8)') 'Entropy = ',entropy
    write(2,'(2x,a16,g16.8)') 'Resulting chi = ',chi2
    write(2,'(2x,a16,g16.8)') 'Test value = ',test
    write(2,'(2x,a16)') '*****************'
    write(5,'(i10,3g16.8)') total,chi2,entropy,test
c
c----- write velocities into log file
c
    do 333 i=1,nbym,10
        write(2,'(10(1x,i5))') (idnint(dble(Image(k))),k=i,i+9)
333 continue
c
c---- stop after one iteration if required
    if(stop.eq. 1) goto 2
    stopc=stopc+1
    if(stop.eq.stopc) goto 2
c
c----- update chi memory vector
c
    do 7000 i=1,4
        chimem(i)=chimem(i+1)
7000 continue
    numtot=numtot+numnull
    chimem(5)=chi2
    write(2,'(2x,a9,i5)')' 'numtot = ', numtot
c-----
    total=total+1
    if(total.gt.maxit) then
        write(2,'(2x,a37)') '->> more than MAXIT iterations in stage'
        if(test.lt.0.2) then
            goto }7
        else
            goto 2
            endif
            endif
c
c---- check to see if chi0 can be set and if stage is finished
c----- check for oscillations and reduce distance limit if necessary
c--.-- check for stability and increase distance limit if necessary
c
    j=0
    k=0
    t1=pctd*chimem(1)
    t2=pctd**chimem(1)
    do }7010\textrm{i}=2,
            if(chimem(i).lt.t1) j=1
            if(chimem(i).lt.t2) k=1
```

```
7 0 1 0
        continue
        x=0
        do 7015 i=2,5
        if((chimem(i).gt.chimem(i-1)).and.(chimem(i-1).ne.0.0)) x=x+1
7 0 1 5
        continue
    if(chimem(1).eq.0.0) then
        j=1
        k=1
        endif
    if(((test.ge.0.2).and.(test.gt.testold)).or.(x.ge.2).or.
    + (chimem(5).gt.chimem(4))) then
        if((dlimit.gt.1000.0).and.(test.lt.0.2)) goto 77
        dlimit=dlimit*2.0
        ucount=ucount+1
        chi0=0.0
        do }7016\quad\textrm{i}=1,
            chimem(i)=chimem(5)
            continue
        chimem(1)=0.0
        testold=test
        iter=iter+1
        goto }
        endif
    if((k.eq.0).and.(test.lt.0.2)) then
        if((L.gt.0.9*L0).and.(ucount.le.10)) then
            dl imit=dl imit*0.667
            testold=test
            iter=iter+1
            goto }
        else
            goto 77
            endif
        endif
    if((x.eq.0).and.(L.gt.0.9*L0).and.(ucount.le.10)) then
        dl imit=dl imit*0.8
        testold=test
        iter=iter+1
        goto }
        endif
C
c----- begin next iteration
    testold=test
    iter=iter+1
    goto }
c
c-\cdots-- reset smoothing stage
77 if(LL.eq.10) goto 2
    total=0
    chi0=0.0
    dlimit=1.0
    testold=0.0
    ucount=0
    LL=LL/2
    if(LL.lt.int(2*mint)) LL=10
    do 7020 i=1,nmax
        Am(i)=Image(i)
7020 continue
    iter=iter+1
    goto }1010
c
c----- else close files and exit
c
2 continue
    close(2)
    close(5)
    close(9)
    close(11)
    stop
    end
```

```
@PROCESS OPT(3),VECTOR(LEVEL(2),REPORT(XLIST)),
@PROCESS FLAG(E),
@PROCESS DIRECTIVE('*VD:')
    SUBROUTINE CMPICK(MEM,LMEM)
    IMPLICIT INTEGER(A-Y)
    DIMENSION MEM(1)
    CALL PICKER (MEM,MEM, LMEM)
    STOP
    END
@PROCESS OPT(3),VECTOR(LEVEL(2),REPORT(XLIST)),
@PROCESS FLAG(E),
@PROCESS DIRECTIVE('*VD:')
    SUBROUTINE PICKER (MEM,ZMEM,LMEM)
    IMPLICIT INTEGER (A-Y)
    COMMON /SCRPTR/ IBSCR,NXTSCR,MAXSCR
    DIMENSION ZPICKS(2000,5)
    DIMENSION ZOFF(150)
    DIMENSION FCB(20),RCB(20),TRACES(2)
    DIMENSION FCB2(20),RCB2(20)
    DIMENSION ACNM(2),ACNM2(2), ZARRAY (200*150),ARRAY(2)
    DIMENSION MEM(1),ZMEM(1)
    DATA ATD /'ATD '/,TRACES/'0001','0001'/
    DATA ACNM/'CRPI','N '/
    DATA ACNM2/'PICK','S1 '/
    DATA GOOD/'GOOD'/,BLNK/' '/
    NXTSCR=1
    MAXSCR=LMEM
    ZTWOPI=8.0*ATAN(1.0)
    ZPI=4.0*ATAN(1.0)
    ZPION2=2.0*ATAN(1.0)
C
    WRITE(6,80000)
80000 FORMAT('1',//' NPCKCRP2 FOR LINE D DATA - O9 JAN 1993',/,
* , ...-.. WITH TRACE HEADER UTILISATION .......,',
* , --------------------------------------------------------
C
C *** OPEN INPUT/OUTPUT FILE
C CALL OPN013(FCB,1,ACNM, 2,0,0,0,4096,IST)
    IF(IST.NE.O)GOTO 90000
C
C *** OPEN INPUT PICKS FILE
C
    CALL OPN013(FCB2,1,ACNM2,1,0,0,0,4096,IST)
    IF(IST.NE.0)GOTO 90001
    CALL DMIO14(ARRAY(1), 2,1,RCB2,FCB2,1,IST)
    MINCRP=ARRAY(1)
    MAXCRP=ARRAY (2)
    WRITE(6,'(1X,A13,2I10)') 'MIN/MAXCRP =',MINCRP,MAXCRP
C
C *** READ OUTPUT FILE AND MAKE SOME CHECKS
C
    IF((NXTSCR+320).GT .MAXSCR)GOTO }9010
    CALL DMIO14(MEM(NXTSCR),320,4096,RCB,FCB,1,IST)
    NXTSCR=NXTSCR-1
    IF(IST.NE.O)GOTO 90010
C
C *** CHECK THAT FILE IS ATD/TRACES
C
    IF(MEM(NXTSCR+1).NE.TRACES(1))GOTO 90020
    IF(MEM(NXTSCR+2).NE.TRACES(2))GOTO 90020
C
C *** SET SDMIN, SDMAX, SDINC,
C *** EDMIN, EDMAX, EDINC, TZERO , TMAX , DELT
C
    SDMIN=MEM(NXTSCR+132)
    NSDMS=MEM(NXTSCR+133)
    SDINC=MEM(NXTSCR+134)
    EDMIN=MEM(NXTSCR+136)
    NEDMS=MEM (NXTSCR+137)
    EDINC=MEM(NXTSCR+138)
```

```
    TZERO=MEM(NXTSCR+152)
    NAMPS=MEM(NXTSCR+153)
    DELT=MEM(NXTSCR+149)
    LTH=MEM(NXTSCR+150)*64
    LTA=MEM(NXTSCR+151)*64
    TRLEN=(LTH+LTA)
    SDMAX=(NSDMS-1)*SDINC+SDMIN
    TMAX=(NAMPS-1)*DELT+TZERO
    LINTV=NSDMS*NEDMS
    SDBHDR=NXTSCR
    NXTSCR=NXTSCR+320
    WRITE(6,'(2X,A10,I10)') 'SDMIN =', SDMIN
    WRITE(6,'(2X,A10,110)') 'SDMAX =',SDMAX
    WRITE(6,'(2X,A10,110)') 'SDINC =',SDINC
    WRITE(6,'(2X,A10,I10)') 'EDMIN =',EDMIN
    WRITE(6,'(2X,A10,110)') 'NEDMS =',NEDMS
    WRITE(6,'(2X,A10,110)') 'EDINC =',EDINC
    WRITE(6,'(2X,A10,110)') 'TZERO =',TZERO
    WRITE(6,'(2X,A10,110)') 'TMAX =',TMAX
    WRITE(6,'(2X,A10,I10)') 'DELT =',DELT
    WRITE(6,'(2X,A10,I10)') 'LTH =', LTH
    WRITE(6,'(2X,A10,I10)') 'LTA =',LTA
    ZDELTT=FLOAT (DELT)
C
C *** GET IV'S
C
    IF((NXTSCR+LINTV).GT.MAXSCR) GOTO 90100
    CALL DMI014(MEM(NXTSCR),LINTV,8192,RCB,FCB,1,IST)
    IF(IST.NE.0)GOTO 90030
C
C *** CHECK ENOUGH ROOM
C
    WRITE(6,'(2X,A10)')
    WRITE(6,'(2X,A10,I10)') 'MAXSCR =',MAXSCR
    WRITE(6,'(2X,A10,I10)') 'NXTSCR =',NXTSCR
    WRITE(6,'(2X,A10,I10)') 'LINTV =',LINTV
    WRITE(6,'(2X,A10,I10)') 'NAMPS =',NAMPS
    IF((NXTSCR+LINTV+3*NAMPS*NEDMS).GT.MAXSCR)
    + GOTO 90100
    TRSCR=NXTSCR+LINTV+LTH
    IF(MOD(FLOAT(TRSCR),2.0).EQ.0.0) TRSCR=TRSCR+1
    WRITE(6,'(2X,A10,I10)') 'TRSCR =',TRSCR
C *** MAIN LOOP
C
    SHTEND=TRSCR+NAMPS*NEDMS
    DO 900 I=1,NSDMS,1
        WRITE(6,'(2X,A18,15)') 'PROCESSING CRP NO.',SDMIN+(I-1)
        DO 905 J=1,NAMPS*NEDMS
            MEM(TRSCR +J-1)=0
905 CONTINUE
C *** READ IN COMPLETE SDOM
    DO 950 J=1,NEDMS, 1
C *** GET IV OF NEXT TRACE
                    K=NXTSCR+(I-1)*NEDMS+(J-1)
                    IV=MEM(K)
                    IF(IV.LE.O) THEN
                    DO }951\textrm{K}=1\mathrm{ ,NAMPS
                            MEM(TRSCR+(J-1)*NAMPS+(K-1))=0.0
951
                    CONTINUE
                    GOTO }95
                    ENDIF
                            IV=IV+4096
C *** GET TRACE HEADER ONLY
                            CALL DMIO14(MEM(TRSCR-LTH),LTH,IV,RCB, FCB,1,IST)
                            ZOFF(J)=NINT(ZMEM(TRSCR-LTH+16))
C *** GET TRACE DATA ONLY
                            CALL DMIO14(MEM(TRSCR+(J-1)*NAMPS),NAMPS,IV+LTH,
                                    RCB,FCB,1,IST)
            IF(IST.NE.O)GOTO 90050
950 CONTINUE
C
```

```
C----- READ ALL THE PICKS FOR THIS CRP
C
    TMP=200+(SDMIN+(I - 1)-MINCRP)*150*200
    CALL DMI014(ZARRAY(1),150*200,TMP,RCB2,FCB2,1,IST)
    IF(IST.NE.O)GOTO 90150
C
C-...- LOOP THROUGH EACH OF THE PICKS
C
    NUMPCK=0
        DO 100 L=1,(NEDMS-1)*200,4
            ZXSRC=ZARRAY(L)
            ZXRCV=ZARRAY (L+1)
            ZTIME=ZARRAY(L+2)
            ZANGRC=ZARRAY(L+3)
            IF(ZXSRC.LT.0) GOTO }10
            IF(ABS(ZXSRC-ZXRCV).LT.50.0) GOTO 100
            IF((ZTIME.LE.0.0).OR.(ZTIME.GT.3.0)) GOTO 100
            CRPTR=INT (FLOAT (L-1)/200.0)+1
            NUMPCK=NUMPCK+1
            IF((CRPTR.EQ.1).OR.(CRPTR.EQ.NEDMS)) GOTO 100
c***** COMPUTE PHASE VELOCITY
            K=(CRPTR-1)*NAMPS
            KN=((CRPTR-1)-1)*NAMPS
            KP=((CRPTR+1)-1)*NAMPS
            J=NINT((ZTIME*1000.0)/ZDELTT)
            ZA=ZMEM(TRSCR +KN+J-1)
            ZAP=ZMEM(TRSCR+KN+J-2)
            ZB=ZMEM(TRSCR+KN+J)
            ZC=ZMEM(TRSCR +KN+J+1)
            ZCP}=\textrm{ZMEM(TRSCR}+\textrm{KN}+\textrm{J}+2
            ZD=ZMEM(TRSCR +K +J-1)
            ZE=ZMEM(TRSCR+K+J)
            ZF=ZMEM(TRSCR +K+J+1)
            ZH=ZMEM(TRSCR+KP+J-1)
            ZHP=ZMEM(TRSCR +KP+J-2)
            ZI =ZMEM(TRSCR+KP+J)
            ZJ=ZMEM(TRSCR+KP+J+1)
            ZJP=ZMEM(TRSCR +KP + J +2)
C
C--.-- CALCULATE THE TIMES OF THE MAXIMA (QUADRATIC INTERP.)
C
C
C----- COMPUTE FIRST TRIPLET TIME (TRACE LESS THAN TARGET TRACE)
C--.-- IF THE ADJACENT TRIPLET DOES NOT SURROUND A MAXIMUM
C----- CHECK THE BEST TRIPLET ONE UP OR ONE DOWN
C
IF((ZA.GT.ZB).OR.(ZC.GT.ZB)) THEN
    IF(ZA.GT.ZC) THEN
                ZT1=ZDELTT*((ZAP-ZB)/(2*(ZAP+ZB-2*ZA)))+(J-2)*ZDELTT
    ELSE
                ZT1=ZDELTT*((ZB-ZCP)/(2*(ZB+ZCP-2*ZC)))+J*ZDELTT
                ENDIF
ELSE
    ZT1=ZDELTT*((ZA-ZC)/(2*(ZA+ZC-2*ZB)))+(J-1)*ZDELTT
    ENDIF
C
C---- COMPUTE FIRST TRIPLET TIME (TRACE GREATER THAN TARGET TRACE)
C----- IF THE ADJACENT TRIPLET DOES NOT SURROUND A MAXIMUM
C----- CHECK THE BEST TRIPLET ONE UP OR ONE DOWN
C
IF((ZH.GT.ZI).OR.(ZJ.GT.ZI)) THEN
    IF(ZH.GT.ZJ) THEN
        ZT3=ZDELTT*((ZHP-ZI)/(2*(ZHP+ZI-2*ZH)))+(J-2)*ZDELTT
    ELSE
                ZT3=ZDELTT*((ZI-ZJP)/(2*(ZI+Z.JP-2*ZJ)))+J*ZDELTT
                ENDIF
ELSE
    ZT3=ZDELTT*}((ZH-ZJ)/(2*(ZH+ZJ-2*ZI)))+(J-1)*ZDELTT
    ENDIF
```

C

```
C----- COMPUTE THE TWO DIPS AND COMPARE (COMPUTE AS VELOCITIES)
C
                    IF(ABS((ZT3-ZT2)-(ZT2-ZT1)).GT.4.0) THEN
                    GOTO 100
                    ENDIF
                    ZV1=(ZOFF(CRPTR+1)-ZOFF(CRPTR))/(ZT3-ZT2)
                    ZV2=(ZOFF(CRPTR)-ZOFF(CRPTR-1))/(ZT2-ZT1)
C
C
C
            ZANGLE=(ZV1+ZV2)/2.0
                    IF(ABS(ZANGLE).LT.1.50) THEN
                    GOTO 100
                    ENDIF
                    ZANGLE=ASIN(1.50/ZANGLE)
                    WRITE(5,6000) ZXSRC,ZXRCV,ZANGLE,ZANGRC, ZTIME
                CONTINUE
        WRITE (6,'(2X,A25,I10)') 'TOTAL PICKS CONSIDERED -',NUMPCK
9 0 0 ~ C O N T I N U E ~
5000 FORMAT (1X, 2G16.8,2I6, 2G16.8)
6000 FORMAT (5G15.6)
C
C *** ClOSE INPUT FILE
C
        CALL CLO074(FCB,0,1,IST)
        IF(IST.NE.0)GOTO 90070
        CLOSE(4)
        ClOSE(5)
C
C
    WRITE(6,80010)
80010 FORMAT(' ',//,' NORMAL COMPLETION')
    STOP
C
C
90000 WRITE(6,90005)
90005 FORMAT(' ',//,' PICKER - ERROR OPENING TRACES FILE',/,
    GOTO }9999
C
90001 WRITE (6,90006)
90006 FORMAT(' ',//,' PICKER - ERROR OPENING PICKS1 FILE',/,
    GOTO 99999
C
9 0 0 1 0 ~ W R I T E ( 6 , 9 0 0 1 5 ) ~
90015 FORMAT('',//,' PICKER -ERROR READING SUBFILE HEADER')
        GOTO 99999
C
90020 WRITE(6,90025)
90025 FORMAT(' ',//,' PICKER - ERROR INPUT FILE NOT ATD/TRACES')
    GOTO 99999
C
90030 WRITE (6,90035)
90035 FORMAT (' ',/l'' PICKER - ERROR READING INTERVAL VECTORS')
        GOTO 99999
C
90040 WRITE (6,90045)ERRCNT
90045 FORMAT(', ///,' PICKER - ERROR OUTPUTTING CRP NO. ',I8)
    GOTO 99999
C
90050 WRITE(6,90055)
90055 FORMAT(' ',//,' PICKER - ERROR READING TRACE DATA')
        GOTO 99999
C
90070 WRITE (6,90075)
90075 FORMAT(' ',//,' PICKER - ERROR CLOSING OUTPUT FILE')
        GOTO }9999
C
90100 WRITE (6,90105)
```

```
90105 FORMAT(' ',//,' PICKER - ERROR - LTP PARAMETER TOO SMALL')
    Gото 99999
C
90150 WRITE (6,90155)
90155 FORMAT(' ',//,' PICKER - ERROR READING PICKS1 DATA')
    GOTO 99999
99999 STOP 15
    END
```

```
@PROCESS OPT(3),VECTOR(LEVEL(2),REPORT(XLIST)),
QPROCESS FLAG(E),
@PROCESS DIRECTIVE('*VD:')
    SUBROUTINE CMPICK(MEM,LMEM)
    IMPLICIT INTEGER(A-Y)
    DIMENSION MEM(1)
    CALL PICKER(MEM,MEM,LMEM)
    STOP
    END
@PROCESS OPT(3),VECTOR(LEVEL(2),REPORT(XLIST)),
@PROCESS FLAG(E),
@PROCESS DIRECTIVE('*VD:')
    SUBROUTINE PICKER (MEM,ZMEM,LMEM)
    IMPLICIT INTEGER (A-Y)
    COMMON /SCRPTR/ IBSCR,NXTSCR,MAXSCR
    COMMON /CMPPTR/ TRSCR,RADIX,SHTEND,NEDMS,NAMPS,AMPSTR,SPC,ZDELTT,
    +
    COMMON /CRPOFF/ CRP,CRPTR,ZOFF
    COMMON /PICKS1/ FCB2,RCB2,NEXT,MINCRP,MAXCRP,NUMPCK
    DIMENSION FCB(20),RCB(20),TRACES(2)
    DIMENSION FCB2(20),RCB2(20)
    DIMENSION ACNM(2),ACNM2(2)
    DIMENSION MEM(1),ZMEM(1)
    DIMENSION CRP(150),CRPTR(150),ZOFF(150)
    DIMENSION NEXT(2000,150)
    DATA ATD /'ATD '/,TRACES/'0001','0001'/
    DATA ACNM/'INPU','T '/
    DATA ACNM2/'PICK','S1 '/
    DATA GOOD/'GOOD'/,BLNK/' '/
    NXTSCR=1
    MAXSCR=LMEM
    MINCRP=0
    MAXCRP=0
    NUMPCK=0
    DO 16 I=1,2000
        DO 17 J=1,150
            NEXT(1,J)=0
                CONTINUE
16 continue
C WRITE(6,80000)
80000 FORMAT('1',//' NPCKSPC2 FOR LINE D DATA - }11\mathrm{ DEC 1992',/,
```



```
C
c *** OPEN INPUT/OUTPUT FILE
C
    CALL OPNO13(FCB,1,ACNM,2,0,0,0,4096,IST)
    IF(IST.NE.0)GOTO 90000
C
c *** OPEN FILE TO STORE THE PICKS
c
    CALL OPN013(FCB2,1,ACNM2,4,0,0,0,4096,IST)
    IF(IST.NE.O)GOTO 90000
C
c *** read output file and make some checks
C
    IF((NXTSCR+320).GT. MAXSCR)GOTO 90100
    CALL DMI014(MEM(NXTSCR),320,4096,RCB,FCB,1,IST)
    NXTSCR=NXTSCR-1
    IF(IST.NE.O)GOTO 90010
C
C *** CHECK THAT FILE IS ATD/TRACES
C
    IF(MEM(NXTSCR+1).NE.TRACES(1))GOTO 90020
    IF(MEM(NXTSCR+2).NE.TRACES(2))GOTO 90020
C
C *** SET SDMIN, SDMAX, SDINC,
C *** EDMIN, EDMAX, EDINC, TZERO , TMAX , DELT
C
    SDMIN=MEM(NXTSCR+132)
```

```
NSDMS=MEM(NXTSCR+133)
SDINC=MEM(NXTSCR+134)
EDMIN=MEM(NXTSCR+136)
NEDMS=MEM(NXTSCR+137)
EDINC=MEM(NXTSCR+138)
TZERO=MEM(NXTSCR+152)
NAMPS=MEM(NXTSCR+153)
DELT=MEM(NXTSCR+149)
LTH=MEM(NXTSCR+150)*64
LTA=MEM(NXTSCR+151)*64
TRLEN=(LTH+LTA)
SDMAX=(NSDMS-1)*SDINC+SDMIN
TMAX=(NAMPS-1)*DELT+TZERO
INTV=NSDMS*NEDMS
SDBHDR=NXTSCR
NXTSCR=NXTSCR+320
WRITE(6,'(2X,A10,I10)') 'SDMIN =',SDMIN
WRITE(6,'(2X,A10,110)') 'SDMAX
WRITE(6,'(2X,A10,110)') 'SDINC =',SDINC
WRITE(6,'(2X,A10,110)') 'EDMIN =',EDMIN
WRITE(6,'(2X,A10,110)') 'NEDMS =',NEDMS
WRITE(6,'(2X,A10,I10)') 'EDINC =',EDINC
WRITE(6,'(2X,A10,110)') 'TZERO =',TZERO
WRITE(6,'(2X,A10,I10)') 'TMAX =',TMAX
WRITE(6,'(2X,A10,110)') 'DELT =',DELT
WRITE(6,'(2X,A10,110)') 'LTH =',LTH
WRITE(6,'(2X,A10,110)') 'LTA =',LTA
ZDELTT=FLOAT(DELT)
C
C *** WRITE ZERO'S TO THE PICKS FILE
C
CALL ZDA060(FCB2,RCB2,0,(NSDMS*4)*150*200,1,IST)
C
C *** GET IV'S
C
IF((NXTSCR+LINTV).GT.MAXSCR) GOTO 90100
CALL DMI014(MEM(NXTSCR),LINTV,8192,RCB,FCB,1,IST)
IF(IST.NE.0)GOTO 90030
C
c----- COMPUTE THE TEMPORAL RADIX TO USE
c
    RADIX=256
    IF(NAMPS.GT.250) RADIX=512
    IF(NAMPS.GT.500) RADIX=1024
    IF(NAMPS.GT.1000) RADIX=2048
    IF(NAMPS.GT.2000) RADIX=4096
    IF(NAMPS.GT.4000) GOTO }9020
C
C *** CHECK ENOUGH ROOM
C
    WRITE(6,'(2X,A10)') '
    WRITE(6,'(2X,A10,I10)') 'MAXSCR =',MAXSCR
    WRITE(6,'(2X,A10,110)') 'RADIX =', RADIX
    WRITE(6,'(2X,A10,110)') 'NXTSCR =',NXTSCR
    WRITE(6,'(2X,A10,I10)') 'LINTV =',LINTV
    WRITE(6,'(2X,A10,I10)') 'NAMPS =',NAMPS
    IF((NXTSCR+LINTV+2*RADIX*NEDMS+3*NAMPS*NEDMS+80000).GT.MAXSCR)
    + GOTO 90100
    TRSCR=NXTSCR+LINTV+LTH
    IF(MOD(FLOAT(TRSCR),2.0).EQ.0.0) TRSCR=TRSCR+1
    WRITE(6,'(2X,A10,110)') 'TRSCR =',TRSCR
C *** MAIN LOOP
C
    SHTEND=TRSCR+2*RADIX*NEDMS
    DO 900 I=1,NSDMS,1
        WRITE(6,'(2X,A18,I5)') 'PROCESSING SPC NO.',SDMIN+(I-1)
        DO 905 J=1,2*RADIX*NEDMS
            MEM(TRSCR +J-1)=0
C05 CONTINUE
    DO 950 J=1,NEDMS,1
```

```
C *** GET IV OF NEXT TRACE
                K=NXTSCR+(I-1)*NEDMS+(J-1)
                IV=MEM(K)
        IF(IV.LE.O) GOTO }95
        IV=IV+4096
C *** GET TRACE HEADER ONLY
                CALL DMI014(MEM(TRSCR-LTH),LTH,IV,RCB,FCB,1, IST)
        CRP(J)=MEM(TRSCR-LTH+40)
        CRPTR(J)=MEM(TRSCR-LTH+41)
        ZOFF(J)=NINT(ZMEM(TRSCR-LTH+16))
c WRITE(6,'(4I10,G16.8)') I,J,CRP(J),CRPTR(J),ZOFF(J)
C *** GET TRACE DATA ONLY
        CALL DMIO14(MEM(TRSCR+RADIX+(J-1)*2*RADIX),LTA,IV+LTH,
                RCB, FCB, 1, IST )
                IF(IST.NE.0)GOTO 90050
9 5 0 ~ C O N T I N U E ~
C
C----- IF THIS IS THE FIRST SHOT, DETERMINE MIN CRP
    IF(MINCRP.EQ.O) THEN
        IF(CRP(1).LT.CRP(NEDMS)) THEN
                MINCRP=CRP(1)
            ELSE
                MINCRP=CRP(NEDMS)
                ENDIF
            ENDIF
C
C---- ADJUST THE TRACE AMLITUDES TO SIMULATE COMPLEX DATA
C
            DO 960 J=1,NEDMS,1
                DO 970 K=1,RADIX, 1
                    MEM(TRSCR+(2*(K-1))+(J-1)*2*RADIX)=
                                    MEM(TRSCR+RADIX+(J-1)*2*RADIX+(K-1))
                    ZMEM(TRSCR+RADIX+(J-1)*2*RADIX+(K-1))=0.0
                    ZMEM(TRSCR+(2*(K-1)+1)+(J-1)*RADIX)=0.0
970 CONTINUE
960 CONTINUE
C
C----- CALL SUBROUTINE TO DO COMPLEX WORK
            SPC=SDMIN+1-1
            CALL CMPLEX(MEM,MEM,MEM,LMEM)
C
C *** WRITE OUT THE INSTANTANEOUS PHASE SHOT RECORD
            ERRCNT=SDMIN+(I-1)*SDINC
            DO 980 J=1,NEDMS,1
C *** GET IV OF NEXT TRACE
            K=NXTSCR+(I-1)*NEDMS+(J-1)
            IV=MEM(K)
            IF(IV.LE.0) GOTO 980
            IV=IV+4096
C *** WRITE TRACE DATA ONLY
                    CALL DMO015(MEM(AMPSTR+(J-1)*RADIX),NAMPS,IV+LTH,
                                    RCB,FCB,1, IST )
            IF(IST.NE.0)GOTO 90040
980 CONTINUE
C
900 CONTINUE
C
C *** CLOSE OUTPUT FILE
    CALL CLO074(FCB,0,1,IST)
    IF(IST.NE.0)GOTO 90070
    CLOSE(4)
C
    WRITE(6,80010)
80010 FORMAT(' ',//,' NORMAL COMPLETION')
    STOP
C
```

```
90000 WRITE(6,90005)
90005 FORMAT(' ',//,',
        GOTO 99999
C
90001 WRITE(6,90006)
90006 FORMAT(' ',//,'
    GOTO 99999
C
9 0 0 1 0 ~ W R I T E ( 6 , 9 0 0 1 5 ) ~
90015 FORMAT(' ',//,'
        GOTO 99999
C
90020 WRITE (6,90025)
90025 FORMAT(' ',//,' PICKER - ERROR INPUT FILE NOT ATD/TRACES')
        GOTO 99999
C
90030 WRITE (6,90035)
90035 FORMAT(', //,' PICKER - ERROR READING INTERVAL VECTORS')
        GOTO 99999
C
90040 WRITE(6,90045)ERRCNT
90045 FORMAT(' ',//,' PICKER - ERROR OUTPUTTING SPC NO. ',18)
        GOTO 99999
C
90050 WRITE(6,90055)
90055 FORMAT(' ',//,' PICKER - ERROR READING TRACE DATA')
        GOTO 99999
C
90070 WRITE (6,90075)
90075 FORMAT(' ',//,' PICKER - ERROR CLOSING OUTPUT FILE')
        GOTO 99999
C
90100 WRITE(6,90105)
90105 FORMAT(' ',//,' PICKER - ERROR - LTP PARAMETER TOO SMALL')
        GOTO 99999
C
90200 WRITE (6,90205)
90205 FORMAT (' ',//,' PICKER - MORE THAN 4000 SAMPLES/TRACE ')
        GOTO 99999
C
90300 WRITE (6,90305)
90305 FORMAT(' ',/l,' PICKER - MORE THAN 490 TRACES/SDOM ')
        GOTO 99999
C
99999 STOP }1
    END
@PROCESS OPT(3),VECTOR(LEVEL(2),REPORT(XLIST)),
@PROCESS FLAG(E),
QPROCESS DIRECTIVE('*VD:')
        SUBROUTINE CMPLEX (MEM, YMEM,ZMEM, LMEM)
        IMPLICIT INTEGER (A-X)
        IMPLICIT COMPLEX*8 (Y)
        COMMON /SCRPTR/ IBSCR,NXTSCR,MAXSCR
        COMMON /CMPPTR/ TRSCR,RADIX,SHTEND,NEDMS,NAMPS,AMPSTR,SPC,ZDELTT,
        + CRP1,STRT, INC,RECNUM
        COMMON /CRPOFF/ CRP,CRPTR,ZOFF
        COMMON /PICKS1/ FCB2,RCB2,NEXT,MINCRP,MAXCRP,NUMPCK
        DIMENSION FCB2(20),RCB2(20)
        DIMENSION MEM(1),YMEM(1),ZMEM(1)
        DIMENSION ZPICKS(150,50,6)
        DIMENSION CRP(150),CRPTR(150),ZOFF(150)
        DIMENSION NEXT(2000,150),ZARRAY(4),ARRAY(2)
        NUMPCK=0
        ZTWOPI=8.0*ATAN(1.0)
        ZPI=4.0*ATAN(1.0)
        ZPION2=2.0*ATAN(1.0)
        ZPION8=0.5*ATAN(1.0)
```

c
c-
PERFORM FORWARD FOURIER TRANSFORMS
c
TRSCR2=(TRSCR +1 ) $/ 2$
SHTEND2=SHTEND/2+2
IF ((SHTEND+120000).GT.MAXSCR) GOTO 90100
CALL SCFT(1, YMEM(TRSCR2), 1, RADIX, YMEM(TRSCR2), 1, RADIX,
+
+
RADIX, NEDMS, 1, 1.0, YMEM (SHTEND2), 20000,
$+\quad$ YMEM(SHTEND2+20000),20000)
CALL SCFT(0, YMEM (TRSCR2) , 1, RADIX, YMEM (TRSCR2), 1,RADIX,

+ RADIX, NEDMS, 1,1.0, YMEM(SHTEND2), 20000,
$+$ YMEM(SHTEND $2+20000$ ), 20000)
C
C--..- ZERO THE NEGATIVE FREQUENCIES
C
DO $100 \mathrm{I}=1$, NEDMS, 1 $Z A=T R S C R 2+(I-1) * R A D I X$ DO $200 \mathrm{~J}=$ RADIX $/ 2+1$, RADIX $\operatorname{YMEM}(Z A+J)=\operatorname{CMPLX}(0.0,0.0)$ CONTINUE
YMEM $(Z A)=Y M E M(Z A) / 2.0$
CONTINUE
c
c
c
CALL $\operatorname{SCFT}(1$, YMEM (TRSCR2) , 1, RADIX, YMEM (TRSCR2) , 1, RADIX,
$+$
$+\quad$ YMEM(SHTEND2+20000),20000)
CALL SCFT( 0 , YMEM(TRSCR2), 1, RADIX, YMEM (TRSCR2), 1, RADIX,
+ RADIX,NEDMS $-1,1.0 /$ FLOAT (RADIX), YMEM (SHTEND2), 20000, YMEM(SHTEND2+20000), 20000)
c
c----- COMPUTE INSTANTANEOUS AMPLITUDE
c
AMPSTR $=$ SHTEND +80100
IF ((AMPSTR+NEDMS*RADIX).GT.MAXSCR) GOTO 90100
Z1 $=0.0$
DO $300 \mathrm{I}=1$, NEDMS
$K=(I-1) * R A D I X$
DO $400 \mathrm{~J}=0$,NAMPS-1
ZMEM $($ AMPSTR $+K+J)=\operatorname{SQRT}($ REAL $($ YMEM $(T R S C R 2+K+J)) * * 2+$
IMAG (YMEM (TRSCR $2+K+J)$ )**2)
IF (ZMEM (AMPSTR $+K+J$ ) $\cdot G T . Z 1$ ) $\quad Z 1=Z M E M(A M P S T R+K+J)$
CONTINUE
CONTINUE
300
c
FIND MAXIMUM AMPLITUDE POINTS
c
VELSTR=AMPSTR+NEDMS*RADIX
IF ((VELSTR+NEDMS*RADIX) . GT .MAXSCR) GOTO 90100
DO $700 \mathrm{I}=2$, NEDMS $-1,1$
$\mathrm{K}=(\mathrm{I}-1)$ *RADIX
$\mathrm{KN}=((\mathrm{I}-1)-1)$ *RADIX
$\mathrm{KP}=((\mathrm{I}+1)-1)$ *RADIX
DO $800 \mathrm{~J}=1$, NAMPS $-2,1$
IF (ZMEM (AMPSTR $+K+J$ ).LT 0.01 *Z1) THEN
ZMEM (VELSTR $+K+J)=0.0$
GOTO 800
ENDIF
IF ( $($ ZMEM (AMPSTR $+K+J) \cdot L T . Z M E M(A M P S T R+K+J-1))$. OR .
(ZMEM (AMPSTR $+K+j$ ) $\cdot L T$. $Z$ MEM (AMPSTR $+K+j+1$ ))) THEN
ZMEM (VELSTR $+K+J)=0.0$
GOTO 800
ENDIF
C---- PICK LARGEST AMPLITUDE WITHIN 200 MSEC ONLY
DO $850 \mathrm{JJ}=-12,-1,1$
IF (( $\mathrm{J}+\mathrm{JJ})$.LT.0) GOTO 850
IF ( $(Z M E M$ (VELSTR $+K+J+J J)$. NE 0.0 ). AND.
(ZMEM (AMPSTR $+K+j+J J)$.LE. ZMEM (AMPSTR $+K+j))$ ) THEN
ZMEM $($ VELSTR $+K+J+J J)=0.0$
ELSEIF ( $(Z M E M(V E L S T R+K+J+J J)$. NE $\cdot 0 \cdot 0)$. AND.
(ZMEM(AMPSTR $+K+J+J J)$.GT.ZMEM (AMPSTR $+K+J$ ))) THEN
ZMEM (VELSTR $+K+J)=0.0$
GOTO 800
ENDIF
$Z A=Z M E M(A M P S T R+K N+J-1)$
$Z A P=Z M E M(A M P S T R+K N+J-2)$ $Z B=Z M E M(A M P S T R+K N+J)$
$\mathrm{ZC}=\mathrm{ZMEM}($ AMPSTR $+K N+J+1)$
ZCP $=$ ZMEM (AMPSTR $+K N+J+2$ )
$Z D=Z M E M(A M P S T R+K+J-1)$
ZE=ZMEM (AMPSTR $+K+J$ )
$Z F=Z M E M($ AMPSTR $+K+J+1)$
$\mathrm{ZH}=\mathrm{ZMEM}($ AMPSTR $+K \mathrm{KP}+\mathrm{J}-1)$
$2 H P=Z M E M(A M P S T R+K P+J-2)$
$\mathrm{ZI}=\mathrm{ZMEM}($ AMPSTR $+K P+J)$
$2 \mathrm{~J}=\mathrm{ZMEM}$ (AMPSTR $+K P+J+1$ )
$Z J P=Z M E M($ AMPSTR $+K P+J+2)$
C
C---- CALCULATE THE TIMES OF THE MAXIMA (QUADRATIC INTERP.)
C----- FOR THE CENTRAL TARGET TRACE
C
ZT2=ZDELTT*((ZD-ZF)/(2*(ZD+ZF-2*ZE)))+(J-1)*ZDELTT
C
C.-.-- COMPUTE FIRST TRIPLET TIME (TRACE LESS THAN TARGET TRACE)

C----- IF THE ADJACENT TRIPLET DOES NOT SURROUND A MAXIMUM
c----- CHECK THE BEST TRIPLET ONE UP OR ONE DOWN
C
IF((ZA.GT. ZB).OR. (ZC.GT.ZB)) THEN
IF (ZA.GT.ZC) THEN
$Z T 1=Z D E L T T^{*}\left((Z A P-Z B) /\left(2^{*}\left(Z A P+Z B-2^{*} Z A\right)\right)\right)+(J-2) * Z D E L T T$
ELSE
$Z T 1=Z D E L T T^{*}\left((Z B-Z C P) /\left(2^{*}(Z B+Z C P-2 * Z C)\right)\right)+J * Z D E L T T$ ENDIF
ELSE
$Z T 1=Z D E L T T^{*}\left((Z A-Z C) /\left(2^{*}(Z A+Z C-2 * Z B)\right)\right)+(J-1) * Z D E L T T$ ENDIF
C
C---- COMPUTE FIRST TRIPLET TIME (TRACE GREATER THAN TARGET TRACE)
C---- IF THE ADJACENT TRIPLET DOES NOT SURROUND A MAXIMUM
C----- CHECK THE BEST TRIPLET ONE UP OR ONE DOWN
C
IF((ZH.GT.ZI).OR.(ZJ.GT.ZI)) THEN
IF (ZH.GT.Z.J) THEN
$Z T 3=Z D E L T T^{*}\left((Z H P-Z I) /\left(2^{*}(Z H P+Z I-2 * Z H)\right)\right)+(J-2) * Z D E L T T$
ELSE
ZT3 $=$ ZDELTT* $\left((Z I-Z J P) /\left(2^{*}(Z I+Z J P-2 * Z J)\right)\right)+J * Z D E L T T$ ENDIF
ELSE
ZT3=ZDELTT* $\left((Z H-Z J) /\left(2^{*}(Z H+Z J-2 * Z I)\right)\right)+(J-1) * Z D E L T T$
ENDIF

## C

C-... COMPARE THE TIMES OF THE MAXIMA
C
IF ((ABS (ZT1-ZT2) .GT. 14.0).OR. (ABS (ZT2-ZT3).GT. 14.0)) THEN
ZMEM (VELSTR $+K+J)=0.0$
GOTO 800
ENDIF
C
C C-... COMPUTE THE TWO DIPS AND COMPARE (COMPUTE AS VELOCITIES)
C
IF (ABS ( $(2 T 3-2 T 2)-(Z T 2-Z T 1)) . G T .4 .0)$ THEN
ZMEM (VELSTR $+K+J)=0.0$
GOTO 800
ENDIF
ZV1 $=($ ZOFF $(I+1)-Z O F F(I)) /(Z T 3-Z T 2)$
ZV2 $=(\operatorname{ZOFF}(1)-Z O F F(I-1)) /(Z T 2-Z T 1)$
C
C-...- AVERAGE VELOCITIES, CONVERT TO ANGLES AND SAVE C

ZMEM $($ VELSTR $+K+J)=(Z V 1+Z V 2) / 2.0$

```
            IF(ABS (ZMEM(VELSTR+K+J)).LT.1 50) THEN
            ZMEM(VELSTR+K+J)=0.0
            GOTO }80
            ENDIF
            ZARRAY(4)=ASIN(1.50/ZMEM(VELSTR+K+J))
            ZARRAY (1)=(401-SPC)*25.0
            ZARRAY(3)=J*0.004
            ZARRAY(2)=ZARRAY(1)+ZOFF(I)
            IF(ZXRCV.GT.10000.0) GOTO }80
C
            TMP=NEXT (CRP(I)-MINCRP,CRPTR(I))
            IF(TMP.GT. 196) THEN
C
                    WRITE(6,'(1X,A22)') 'MAX PICKS/TR EXCEEDED'
                    GOTO 800
                    END IF
            TMP=TMP+200+(CRP (I)-MINCRP)*150*200+(CRPTR(I)-1)*200
            CALL DM0015(ZARRAY(1),4,TMP,RCB2,FCB2,1,IST)
            IF(IST.NE.O) GOTO }9011
            NEXT(CRP(I)-MINCRP,CRPTR(I))=NEXT(CRP(I)-MINCRP,CRPTR(I))+4
            IF(CRP(I).GT . MAXCRP) MAXCRP=CRP(I)
            NUMPCK=NUMPCK+1
            CONTINUE
                CONTINUE
    ARRAY(1)=MINCRP
    ARRAY(2)=MAXCRP
    WRITE(6,'(2X,A12,I10)') 'NUMPCK =' ,NUMPCK
    WRITE(6,'(2X,A12,2110)') 'MIN/MAXCRP =' ,MINCRP,MAXCRP
    CALL DMO015(ARRAY(1),2,1,RCB2,FCB2,1,IST)
    IF(IST.NE.0) GOTO }9012
    RETURN
9 0 1 0 0 ~ W R I T E ( 6 , 9 0 1 0 5 ) ~
90105 FORMAT(' ',//,'
                    CMPLEX - ERROR - LTP PARAMETER TOO SMALL')
        GOTO 99999
C
9 0 1 0 1 ~ W R I T E ~ ( 6 , 9 0 1 0 6 ) ~
90106 FORMAT(' ',//,' CMPLEX - ERROR - TOO MANY PICKS ON TRACE')
        GOTO }9999
C
9 0 1 1 0 ~ W R I T E ( 6 , 9 0 1 1 5 )
90115 FORMAT(', ,//,' CMPLEX - ERROR - WRITING TO PICKS1 ')
    GOTO 99999
C
90120 WRITE(6,90125)
90125 FORMAT(' ',//,' CMPLEX - ERROR - WRITING PICKS1 HEADER ')
        GOTO 99999
C
9 9 9 9 9 ~ S T O P ~ 1 5 ~
    END
```

```
@PROCESS OPT(3),VECTOR(LEVEL(2),REPORT(XLIST)),
@PROCESS FLAG(E),
@PROCESS DIRECTIVE('*VD:')
    SUBROUTINE CMPICK(MEM,LMEM)
    IMPLICIT INTEGER(A-Y)
    DIMENSION MEM(1)
    CALL PICKER(MEM,MEM,LMEM)
    STOP
    END
@PROCESS OPT(3),VECTOR(LEVEL(2),REPORT(XLIST)),
@PROCESS FLAG(E),
@PROCESS DIRECTIVE('*VD:')
    SUBROUTINE PICKER (MEM,ZMEM,LMEM)
    IMPLICIT INTEGER (A-Y)
    COMMON /SCRPTR/ IBSCR,NXTSCR,MAXSCR
    DIMENSION ZPICKS(2000,5)
    DIMENSION ZOFF(150)
    DIMENSION FCB(20),RCB(20),TRACES(2)
    DIMENSION ACNM(2)
    DIMENSION MEM(1),ZMEM(1)
    DATA ATD /'ATD '/,TRACES/'0001','0001'/
    DATA ACNM/'CRPI','N '/
    DATA GOOD/'GOOD'/,BLNK/' '/
    NXTSCR=1
    MAXSCR=LMEM
    ZTWOPI=8.0*ATAN(1.0)
    ZPI=4.0*ATAN(1.0)
    ZPION2=2.0*ATAN(1.0)
c
    WRITE(6,80000)
80000 FORMAT('1',//' PICKCRP5 FOR LINE C DATA - }15\mathrm{ AUG 1992',/,
```



```
C
c *** OPEN INPUT/OUTPUT FILE
c
    CALL OPN013(FCB,1,ACNM,2,0,0,0,4096,IST)
    IF(IST.NE.0)GOTO 90000
C
C *** READ OUTPUT FILE AND MAKE SOME CHECKS
c
    IF((NXTSCR+320).GT.MAXSCR)GOTO 90100
    CALL DMI014(MEM(NXTSCR),320,4096,RCB,FCB,1,IST)
    NXTSCR=NXTSCR-1
    IF(IST.NE.0)GOTO 90010
C
c *** CHECK THAT FILE IS ATD/TRACES
C
    IF(MEM(NXTSCR+1).NE.TRACES(1))GOTO 90020
    IF(MEM(NXTSCR+2) .NE.TRACES(2))GOTO 90020
C
C *** SET SDMIN, SDMAX, SDINC,
c *** EDMIN, EDMAX, EDINC, TZERO , TMAX , DELT
C
    SDMIN=MEM(NXTSCR+132)
    NSDMS=MEM(NXTSCR+133)
    SDINC=MEM(NXTSCR+134)
    EDMIN=MEM(NXTSCR+136)
    NEDMS=MEM(NXTSCR+137)
    EDINC=MEM(NXTSCR+138)
    TZERO=MEM(NXTSCR+152)
    NAMPS=MEM(NXTSCR+153)
    DELT=MEM(NXTSCR+149)
    LTH=MEM(NXTSCR+150)*64
    LTA=MEM(NXTSCR+151)*64
    TRLEN=(LTH+LTA)
    SDMAX=(NSDMS-1)*SDINC+SDMIN
    TMAX=(NAMPS-1)*DELT+TZERO
    LINTV=NSDMS*NEDMS
    SDBHDR=NXTSCR
    NXTSCR=NXTSCR+320
```

```
    WRITE(6,'(2X,A10,I10)') 'SDMIN =',SDMIN
    WRITE(6,'(2X,A10,I10)') 'SDMAX =', SDMAX
    WRITE(6,'(2X,A10,I10)') 'SDINC =',SDINC
    WRITE(6,'(2X,A10,I10)') 'EDMIN =',EDMIN
    WRITE(6,'(2X,A10,I10)') 'NEDMS =',NEDMS
    WRITE(6,'(2X,A10,I10)') 'EDINC =',EDINC
    WRITE(6,'(2X,A10,I10)') 'TZERO =',TZERO
    WRITE(6,'(2X,A10,110)') 'TMAX =',TMAX
    WRITE(6,'(2X,A10,110)') 'DELT =',DELT
    WRITE(6,'(2X,A10,110)') 'LTH =',LTH
    WRITE(6,'(2X,A10,I10)') 'LTA =',LTA
    ZDELTT=FLOAT (DELT)
C
C
    IF((NXTSCR+LINTV).GT .MAXSCR) GOTO 90100
    CALL DMIO14(MEM(NXTSCR),LINTV,8192,RCB,FCB,1,IST)
    IF(IST.NE.O)GOTO 90030
C
C *** CHECK ENOUGH ROOM
C
    WRITE(6,'(2X,A10)')', 'MAXSCR =',MAXSCR
    WRITE(6,'(2X,A10,I10)') 'MAXSSCR =',MAXSCR
    WRITE(6,'(2X,A10,I10)') 'LINTV =',LINTV
    WRITE(6,'(2X,A10,I10)') 'NAMPS =',NAMPS
    IF((NXTSCR+LINTV+3*NAMPS*NEDMS).GT .MAXSCR)
    + GOTO 90100
    TRSCR=NXTSCR+LINTV+LTH
    IF(MOD(FLOAT(TRSCR),2.0).EQ.0.0) TRSCR=TRSCR+1
    WRITE(6,'(2X,A10,110)') 'TRSCR =',TRSCR
C *** MAIN LOOP
C
    SHTEND=TRSCR+NAMPS*NEDMS
    DO 900 I=1,NSDMS,1
        DO 1200 J=1,2000
            ZPICKS(J,5)=0.0
            CONTINUE
            WRITE(6,'(2X,A18,15)') 'PROCESSING CRP NO.',SDMIN+(I-1)
            DO 905 J=1,NAMPS*NEDMS
            MEM(TRSCR +J-1)=0
905 CONTINUE
C *** READ IN COMPLETE SDOM
            DO 950 J=1,NEDMS,1
C *** GET IV OF NEXT TRACE
            K=NXTSCR+(I-1)*NEDMS+(J-1)
            IV=MEM(K)
            IF(IV.LE.O) THEN
                    DO 951 K=1,NAMPS
                                    MEM(TRSCR+(J-1)*NAMPS+(K-1))}=0.
9 5 1
                                    CONTINUE
                                    GOTO }95
                    ENDIF
            IV=IV+4096
C *** GET TRACE HEADER ONLY
                            CALL DMI014(MEM(TRSCR-LTH),LTH,IV,RCB,FCB,1,IST)
                            ZOFF(J)=NINT (ZMEM(TRSCR-LTH+16))
C *** GET TRACE DATA ONLY
            CALL DMIO14(MEM(TRSCR+(J-1)*NAMPS),NAMPS,IV+LTH,
                    RCB, FCB, 1, IST )
            IF(IST.NE.0)GOTO 90050
9 5 0 ~ C O N T I N U E ~
C
            COUNT=1
            REWIND (4)
READ(4,5000,END=1000) ZXSRC,ZTIME,CRP,CRPTR,ZXRCV,ZANGRC
            IF(CRP.EQ.SDMIN+(I-1)) THEN
                        IF((CRPTR.EQ.1).OR.(CRPTR.EQ.NEDMS)) GOTO 100
c***** COMPUTE PHASE VELOCITY
```

$\mathrm{K}=($ CRPTR -1$)$ *NAMPS
KN $=(($ CRPTR -1$)-1)$ *NAMPS
KP=((CRPTR+1)-1)*NAMPS
$J=$ INT ( $(2 T I M E * 1000.0) /$ ZDELTT $)+1$ $\operatorname{WRITE}\left(6,^{\prime}(2 X, A 3,1 X, 110)^{\prime}\right)^{\prime}-\mathrm{J}-1, \mathrm{~J}$
$Z A=Z M E M($ TRSTR $+K+J+1)$
$Z M=Z M E M(T R S C R+K+J)$
ZB $=$ ZMEM (TRSCR $+\mathrm{K}+\mathrm{J}-1$ )
$Z \mathrm{C}=\mathrm{ZMEM}($ TRSCR $+\mathrm{KP}+\mathrm{J})$
ZD $=$ ZMEM (TRSCR $+K N+J$ )
C----- CALCULATE DIFFERENCES
CALL DIFF (ZA,ZM,ZU1)
CALL DIFF (ZM, ZB,ZU2)
CALL DIFF (ZC, ZM, ZL1)
CALL DIFF (ZM, ZD, ZL2)
C----- IF TWO HALF DIFFERENCES ARE VERY DIFFERENT, IGNORE POINT
ZT=ZU1/ZU2
IF((ZT.LT.0.25).OR.(ZT.GT.4.00)) GOTO 100
ZT=ZL1/ZL2
IF((ZT.LT.0.25).OR.(ZT.GT.4.00)) GOTO 100
C----- COMPUTE AVERAGE DIFFERENCES
$\mathrm{ZU}=(\mathrm{ZU} 1+\mathrm{ZU} 2) / 2.0$
ZL=(ZL1+ZL2)/2.0
C-- -- CHECK TO SEE IF DIP IS NEGATIVE AND CORRECT IF SO
IF(ZL.GT.ZPI) ZL=ZL-ZTWOPI
ZPVEL=ZU/ZL
ZDELTX $=$ ABS (ZOFF (CRPTR+1)-ZOFF (CRPTR-1) )/2.0
ZPVEL=-1.0*ZPVEL*ZDELTX/ZDELTT
C***** CONVERT PHASE VELOCITIES TO ANGLES AND WRITE TO OUTPUT FILE
IF (ABS (ZPVEL).LT. 1.48) THEN
GOTO 100
END IF
ZANGSC=ASIN(1.48/ZPVEL)
ZPICKS (COUNT, 1)=ZXSRC
ZPICKS (COUNT, 2) $=$ ZXRCV
ZPICKS (COUNT, 3) $=$ ZANGSC
ZPICKS (COUNT,4)=ZANGRC
ZPICKS (COUNT,5)=2TIME
COUNT $=$ COUNT +1
ENDIF
GOTO 100
C
C---- OUTPUT VALUES
C
1000 DO $1110 \mathrm{~J}=1,2000$
ZTIME=ZPICKS (J,5)
IF(ZTIME.EQ.O.0) GOTO 900
WRITE(5,6000) ZPICKS (J, 1), ZPICKS(J,2),
ZPICKS(J,3),ZPICKS( $\downarrow, 4$ ), ZPICKS $(J, 5)$
1110
CONTINUE
C
900 CONTINUE
5000 FORMAT (1X, 2G16.8,216,2G16.8)
6000 FORMAT (5G15.6)
C
C *** CLOSE INPUT FILE
CALL CLO074(FCB, 0, 1, IST)
IF(IST.NE.0)GOTO 90070
CLOSE (4)
CLOSE(5)
C
C
WRITE (6,80010)
80010 FORMAT(',$/ /$,' NORMAL COMPLETION')
STOP
C
$90000 \operatorname{WRITE}(6,90005)$
90005 FORMAT (' ',/l,' PICKER - ERROR OPENING TRACES FILE' ,/, ENSURE ACNM OF INPUT FILE IS CRPIN')

## C

$90010 \operatorname{WRITE}(6,90015)$
90015 FORMAT (',$/ /$,' PICKER -ERROR READING SUBFILE HEADER') GOTO 99999
C
$90020 \operatorname{WRITE}(6,90025)$
90025 FORMAT (',$/ /$, ' PICKER - ERROR INPUT FILE NOT ATD/TRACES') GOTO 99999

C
90030 WRITE $(6,90035)$
90035 FORMAT (' ',//,' PICKER - ERROR READING INTERVAL VECTORS') GOTO 99999
C
90040 WRITE $(6,90045)$ ERRCNT
90045 FORMAT (',$/ /, '$ PICKER - ERROR OUTPUTTING CRP NO. ' , 18) GOTO 99999
C
90050 WRITE $(6,90055)$
90055 FORMAT(' '//,' PICKER - ERROR READING TRACE DATA') GOTO 99999
C
90070 WRITE $(6,90075)$
90075 FORMAT (',$/ /$,' PICKER - ERROR CLOSING OUTPUT FILE') GOTO 99999

## C

$90100 \operatorname{WRITE}(6,90105)$
90105 FORMAT(' ',//,' PICKER - ERROR - LTP PARAMETER TOO SMALL') GOTO 99999
C
99999 STOP 15
END
SUBROUTINE DIFF(ZA,ZM,ZU1)
REAL ZA, ZM, ZU1, ZTWOPI
ZTWOPI =8.0*ATAN(1.0)
IF (( (ZA.GE. O.0) .AND - (ZM.GE 0.0$))$.OR.
$+((Z A . L T \cdot 0.0)$.AND.(ZM.LT. O.0))) THEN
IF (ZA.GE.ZM) THEN
$Z \cup 1=Z A-Z M$
ELSE
ZU1=ZTWOPI - (ZM-ZA)
ENDIF
ENDIF
IF ((ZA.GE.O.0).AND. (ZM.LT. O.0)) ZU1 $=Z A-Z M$ IF((ZA.LT.O.0).AND. (ZM.GE.0.0)) $\quad Z U 1=Z T W O P I-(Z M-Z A)$ RETURN
END

```
@PROCESS OPT(3),VECTOR(LEVEL(2),REPORT(XLIST)),
@PROCESS FLAG(E),
@PROCESS DIRECTIVE('*VD:')
    SUBROUTINE CMPICK(MEM,LMEM)
    IMPLICIT INTEGER(A-Y)
    DIMENSION MEM(1)
    CALL PICKER (MEM,MEM,LMEM)
    STOP
    END
@PROCESS OPT(3),VECTOR(LEVEL(2),REPORT(XLIST)),
aPROCESS FLAG(E),
aPROCESS DIRECTIVE('*VD:')
    SUBROUTINE PICKER (MEM,ZMEM,LMEM)
    IMPLICIT INTEGER (A-Y)
    COMMON /SCRPTR/ IBSCR,NXTSCR,MAXSCR
    COMMON /CMPPTR/ TRSCR,RADIX,SHTEND,NEDMS,NAMPS,PHSSTR,SPC,ZDELTT
    COMMON /CRPOFF/ CRP,CRPTR,ZOFF
    DIMENSION FCB(20),RCB(20),TRACES(2)
    DIMENSION ACNM(2)
    DIMENSION MEM(1),ZMEM(1)
    DIMENSION CRP(150),CRPTR(150),ZOFF(150)
    DATA ATD /'ATD '/,TRACES/'0001','0001'/
    DATA ACNM/'INPU','T '/
    DATA GOOD/'GOOD'/,BLNK/' '/
    NXTSCR=1
    MAXSCR=LMEM
C
    WRITE(6,80000)
80000 FORMAT('1',
    FORMAT('1',//' PICKSPC5 FOR LINE C DATA - 15 AUG 1992',/,
    * , --.--.------------------------------------------
C
c *** OPEN INPUT/OUTPUT FILE
C
    CALL OPN013(FCB, 1,ACNM,2,0,0,0,4096,1ST)
    IF(IST.NE.O)GOTO 90000
c
c *** read OUTPUT FILE AND MAKE SOME CHECKS
C
    IF((NXTSCR+320).GT.MAXSCR)GOTO 90100
    CALL DMI014(MEM(NXTSCR),320,4096,RCB,FCB,1,IST)
    NXTSCR=NXTSCR-1
    IF(IST.NE.O)GOTO 90010
C
c *** CHECK THAT FILE IS ATD/TRACES
C
    IF(MEM(NXTSCR+1).NE.TRACES(1))GOTO 90020
    IF(MEM(NXTSCR+2).NE.TRACES(2))GOTO 90020
C
c *** SET SDMIN, SDMAX, SDINC,
c *** EDMIN, EDMAX, EDINC, TZERO , TMAX , DELT
c
    SDMIN=MEM(NXTSCR+132)
    NSDMS=MEM(NXTSCR+133)
    SDINC=MEM(NXTSCR+134)
    EDMIN=MEM(NXTSCR+136)
    NEDMS=MEM(NXTSCR+137)
    EDINC=MEM(NXTSCR+138)
    TZERO=MEM(NXTSCR+152)
    NAMPS=MEM(NXTSCR+153)
    DELT=MEM(NXTSCR+149)
    LTH=MEM(NXTSCR+150)*64
    LTA=MEM(NXTSCR+151)*64
    TRLEN=(LTH+LTA)
    SDMAX=(NSDMS-1)*SDINC+SDMIN
    TMAX =(NAMPS-1)*DELT+TZERO
    LINTV=NSDMS*NEDMS
    SDBHDR=NXTSCR
    NXTSCR=NXTSCR+320
    WRITE(6,'(2X,A10,I10)') 'SDMIN =',SDMIN
    WRITE (6,'(2X,A10,110)') 'SDMAX =',SDMAX
```

```
    WRITE(6,'(2X,A10,110)') 'SDINC =', SDINC
    WRITE(6,'(2X,A10,I10)') 'EDMIN =',EDMIN
    WRITE(6,'(2X,A10,110)') 'NEDMS =',NEDMS
    WRITE(6,'(2X,A10,I10)') 'EDINC =',EDINC
    WRITE(6,'(2X,A10,I10)') 'TZERO =',TZERO
    WRITE(6,'(2X,A10,I10)') 'TMAX =',TMAX
    WRITE(6,'(2X,A10,110)') 'DELT =',DELT
    WRITE(6,'(2X,A10,I10)') 'LTH =',LTH
    WRITE(6,'(2X,A10,110)') 'LTA =',LTA
    ZDELTT=FLOAT (DELT)
C
C *** GET IV'S
C
    IF((NXTSCR+LINTV).GT.MAXSCR) GOTO }9010
    CALL DMIO14(MEM(NXTSCR),LINTV,8192,RCB,FCB,1,IST)
    IF(IST.NE.0)GOTO 90030
c
c----- COMPUTE THE TEMPORAL RADIX TO USE
c
    RADIX=256
    IF(NAMPS.GT.250) RADIX=512
    IF(NAMPS.GT.500) RADIX=1024
    IF(NAMPS.GT. 1000) RADIX=2048
    IF(NAMPS.GT.2000) RADIX=4096
    IF(NAMPS.GT.4000) GOTO }9020
C
C *** CHECK ENOUGH ROOM
C
        WRITE(6,'(2X,A10)')',
        WRITE(6,'(2X,A10,I10)') 'RADIX =',RADIX
        WRITE(6,'(2X,A10,110)') 'NXTSCR =',NXTSCR
        WRITE(6,'(2X,A10,I10)') 'LINTV =',LINTV
        WRITE(6,'(2X,A10,I10)') 'NAMPS =',NAMPS
        IF((NXTSCR+LINTV+2*RADIX*NEDMS+3*NAMPS*NEDMS+80000).GT.MAXSCR)
    + GOTO 90100
        TRSCR=NXTSCR+LINTV+LTH
        IF(MOD(FLOAT(TRSCR),2.0).EQ.0.0) TRSCR=TRSCR+1
        WRITE(6,'(2X,A10,I10)')'TRSCR =',TRSCR
C *** MAIN LOOP
C
    SHTEND=TRSCR+2*RADIX*NEDMS
    DO 900 I=1,NSDMS,1
        WRITE(6,'(2X,A18,I5)') 'PROCESSING SPC NO.',SDMIN+(I-1)
        DO 905 J=1,2*RADIX*NEDMS
            MEM(TRSCR +J-1)=0
            CONTINUE
C *** READ IN COMPLETE SDOM
            DO 950 J=1,NEDMS,1
C *** GET IV OF NEXT TRACE
                    K=NXTSCR+(I-1)*NEDMS+(J-1)
                    IV=MEM(K)
                    IF(IV.LE.0) GOTO }95
                    IV=IV+4096
C *** GET TRACE HEADER ONLY
            CALL DMIO14(MEM(TRSCR-LTH),LTH,IV,RCB,FCB,1,IST)
            CRP(J)=MEM(TRSCR -LTH+40)
            CRPTR (J)=MEM (TRSCR-LTH+41)
            ZOFF(J)=NINT(ZMEM(TRSCR-LTH+16))
C *** GET TRACE DATA ONLY
            CALL DMI014(MEM(TRSCR+RADIX+(J-1)*2*RADIX),LTA,IV+LTH,
                        RCB, FCB, 1, IST)
            IF(IST.NE.0)GOTO 90050
            CONTINUE
950
C
C--.-- ADJUST THE TRACE AMLITUDES TO SIMULATE COMPLEX DATA
C
    DO 960 J=1,NEDMS, 1
    DO 970 K=1,RADIX,1
                MEM(TRSCR+(2*(K-1))+(J-1)*2*RADIX)=
                        MEM(TRSCR+RADIX+(J-1)*2*RADIX+(K-1))
```

```
                                    ZMEM(TRSCR+RADIX+(J-1)*2*RADIX+(K-1))=0.0
                                    ZMEM(TRSCR+(2*}(K-1)+1)+(J-1)*RADIX)=0.
970 CONTINUE
960 CONTINUE
C
    CALL SUBROUTINE TO DO COMPLEX WORK
            SPC=SDMIN+I-1
            CALL CMPLEX(MEM,MEM,MEM,LMEM)
C
C *** WRITE OUT THE INSTANTANEOUS PHASE SHOT RECORD
            ERRCNT=SDMIN+(I-1)*SDINC
            DO 980 J=1,NEDMS,1
C *** GET IV OF NEXT TRACE
                    K=NXTSCR+(I-1)*NEDMS+(J-1)
                    IV=MEM(K)
                    IF(IV.LE.O) GOTO }98
                    IV=IV+4096
C *** WRITE TRACE DATA ONLY
                    CALL DM0015(MEM(PHSSTR+(J-1)*RADIX),NAMPS,IV+LTH,
                    RCB,FCB,1,IST)
                            IF(IST.NE.O)GOTO 90040
980 CONTINUE
C
9 0 0 ~ C O N T I N U E ~
C *** CLOSE OUTPUT FILE
    CALL CLO074(FCB,0,1,IST)
    IF(IST.NE.0)GOTO 90070
    CLOSE(4)
C
    WRITE(6,80010)
80010 FORMAT(' ',//,' NORMAL COMPLETION')
    STOP
C
C
90000 WRITE(6,90005)
90005 FORMAT(' ',//,' PICKER - ERROR OPENING TRACES FILE',/,
    ENSURE ACNM OF INPUT FILE IS INPUT')
    GOTO 99999
C
90010 WRITE(6,90015)
90015 FORMAT(' ',//,' PICKER -ERROR READING SUBFILE HEADER')
    GOTO 99999
C
90020 WRITE (6,90025)
90025 FORMAT(' ',//,' PICKER - ERROR INPUT FILE NOT ATD/TRACES')
    GOTO 99999
C
9 0 0 3 0 \operatorname { W R I T E } ( 6 , 9 0 0 3 5 )
90035 FORMAT('',//,' PICKER - ERROR READING INTERVAL VECTORS')
    GOTO 99999
C
90040 WRITE(6,90045)ERRCNT
90045 FORMAT(' ',//,' PICKER - ERROR OUTPUTTING SPC NO. ',18)
        GOTO }9999
C
90050 WRITE (6,90055)
90055 FORMAT(' ',//,' PICKER - ERROR READING TRACE DATA')
    GOTO 99999
C
90070 WRITE(6,90075)
90075 FORMAT(' ',//,' PICKER - ERROR CLOSING OUTPUT FILE')
    GOTO 99999
C
90100 WRITE(6,90105)
90105 FORMAT(' ',//,' PICKER - ERROR - LTP PARAMETER TOO SMALL')
        GOTO 99999
C
```

```
9 0 2 0 0 ~ W R I T E ( 6 , 9 0 2 0 5 )
90205 FORMAT(' ',//,' PICKER - MORE THAN 4000 SAMPLES/TRACE ')
        GOTO }9999
C
90300 WRITE(6,90305)
90305 FORMAT(', '//,' PICKER - MORE THAN 490 TRACES/SDOM ')
    GOTO 99999
C
C
9 9 9 9 9 ~ S T O P ~ 1 5 ~
    END
@PROCESS OPT(3),VECTOR(LEVEL(2),REPORT(XLIST)),
@PROCESS FLAG(E),
@PROCESS DIRECTIVE('*VD:')
    SUBROUTINE CMPLEX (MEM,YMEM,ZMEM, LMEM)
    IMPLICIT INTEGER (A-X)
    IMPLICIT COMPLEX*8 (Y)
    COMMON /SCRPTR/ IBSCR,NXTSCR,MAXSCR
    COMMON /CMPPTR/ TRSCR,RADIX,SHTEND,NEDMS,NAMPS,PHSSTR,SPC, ZDELTT
    COMMON /CRPOFF/ CRP,CRPTR,ZOFF
    DIMENSION FCB(20),RCB(20),TRACES(2)
    DIMENSION MEM(1), YMEM(1),ZMEM(1)
    DIMENSION ZPICKS(150,50,6)
    DIMENSION CRP(150),CRPTR(150),ZOFF(150)
    ZTWOPI=8.0*ATAN(1.0)
    ZPI=4.0*ATAN(1.0)
    ZPION2=2.0*ATAN(1.0)
c
c-\cdots-- PERFORM FORWARD FOURIER TRANSFORMS
c
        TRSCR2=(TRSCR +1)/2
        SHTEND2=SHTEND/2+2
        IF((SHTEND+120000).GT.MAXSCR) GOTO 90100
        CALL SCFT(1, YMEM(TRSCR2),1,RADIX, YMEM(TRSCR2),1,RADIX,
        +
        + YMEM(SHTEND2+20000), 20000)
            CALL SCFT(O, YMEM(TRSCR2), 1,RADIX, YMEM(TRSCR2),1,RADIX,
        + RADIX,NEDMS,1,1.0,YMEM(SHTEND2),20000,
        + YMEM(SHTEND2+20000),20000)
C
C----- ZERO THE NEGATIVE FREQUENCIES
C
    DO 100 I=1,NEDMS,1
                ZA=TRSCR2+(I-1)*RADIX
                    DO 200 J=RADIX/2+1,RADIX
                                    YMEM(ZA+J)=CMPLX(0.0,0.0)
                                    CONTINUE
            YMEM(ZA)=YMEM (ZA)/2.0
    CONTINUE
100
c
C----- PERFORM INVERSE 2D FOURIER TRANSFORM
c
    CALL SCFT(1, YMEM(TRSCR2), 1, RADIX, YMEM(TRSCR2), 1,RADIX,
        + RADIX,NEDMS, -1,1.0/FLOAT(RADIX), YMEM(SHTEND2),20000
        + YMEM(SHTEND2+20000),20000)
            CALL SCFT(0, YMEM(TRSCR2), 1,RADIX, YMEM(TRSCR2), 1, RADIX,
        +
        RADIX,NEDMS,-1,1.0/FLOAT(RADIX),YMEM(SHTEND2),20000
                                YMEM(SHTEND2+20000), 20000)
c
c-\cdots--- COMPUTE INSTANTANEOUS AMPLITUDE
c
    AMPSTR=SHTEND +80100
    IF((AMPSTR+NEDMS*RADIX).GT.MAXSCR) GOTO 90100
    Z1=0.0
    DO 300 I=1,NEDMS
            K=(I-1)*RADIX
            DO 400 J=0,NAMPS-1
                ZMEM(AMPSTR +K+J)=SQRT (REAL (YMEM (TRSCR 2+K+J))**2+
                    IMAG(YMEM(TRSCR2+K+J))**2)
                    IF (ZMEM(AMPSTR+K+J).GT.Z1) Z1=ZMEM(AMPSTR +K+J)
```

```
4 0 0
                            CONTINUE
CONTINUE
3
c
C--.-- COMPUTE INSTANTANEOUS PHASE
c
    PHSSTR=AMPSTR+NEDMS*RADIX
    IF((PHSSTR+NEDMS*RADIX).GT.MAXSCR) GOTO 90100
    DO 500 I=1,NEDMS,1
            K=(I-1)*RADIX
            DO 600 J=1,NAMPS-2,1
                    IF(ABS(REAL(YMEM(TRSCR2+K+J))).LT.1.E-10) THEN
                    IF(IMAG(YMEM(TRSCR2+K+J)).LT.0.0) THEN
                            ZMEM(PHSSTR+K+J)=-1.0*ZPION2
                    ELSE
                    ZMEM(PHSSTR+K+J)=ZPION2
                    ENDIF
                    GOTO }60
                    ENDIF
                    ZA=ATAN(IMAG(YMEM(TRSCR2+K+J))/REAL(YMEM(TRSCR2+K+J)))
                IF(REAL(YMEM(TRSCR2+K+J)).LT.0.0) THEN
                    IF(ZA.LT.O.0) THEN
                            ZA=ZPI+ZA
                    ELSE
                        ZA=-1.0*ZPI+ZA
                        ENDIF
                    ENDIF
            ZMEM(PHSSTR+K+J)=ZA
                CONTINUE
600 CONTIN
500
c
c
    VELSTR=PHSSTR+NEDMS*RADIX
    IF((VELSTR+NEDMS*RADIX).GT.MAXSCR) GOTO 90100
    DO 700 I=2,NEDMS-1,1
        K=(I-1)*RADIX
        KN=((I-1)-1)*RADIX
        KP=((I+1)-1)*RADIX
        DO }800\textrm{J}=1\mathrm{ ,NAMPS-2,1
            IF(ZMEM(AMPSTR+K+J).LT 0.05*Z1) THEN
                    ZMEM(VELSTR+K+J)=0.0
                    GOTO }80
                    ENDIF
            IF((ZMEM(AMPSTR+K+J).LT.ZMEM(AMPSTR+K+J-1)).OR.
                    (ZMEM(AMPSTR +K+J) -LT.ZMEM(AMPSTR+K+J+1))) THEN
                    ZMEM(VELSTR +K+J)=0.0
                    GOTO 800
                    ENDIF
C----- PICK LARGEST AMPLITUDE WITHIN 200 MSEC ONLY
            DO }850\textrm{JJ}=-12,-1,
                    IF((J+JJ).LT.0) GOTO }85
                    F((ZMEM(VELSTR+K+J+JJ).NE,0,0).AND.
                    (ZMEM(AMPSTR+K+J+JJ).LE.ZMEM(AMPSTR+K+J))) THEN
                    ZMEM(VELSTR +K + J +JJ)=0.0
                    ELSEIF((ZMEM(VELSTR+K+J+JJ).NE.O.0).AND.
                    (ZMEM(AMPSTR+K+J+JJ).GT.ZMMM(AMPSTR+K+J))) THEN
                    ZMEM(VELSSTR+K+J)=0.0
                    GOTO 800
                    ENDIF
850
                    CONTINUE
            ZA=ZMEM(PHSSTR +K+J+1)
            ZM=ZMEM(PHSSTR+K+J)
            ZB=ZMEM(PHSSTR +K+J-1)
            ZC=ZMEM(PHSSTR +KP +J)
            ZD=ZMEM(PHSSTR+KN+J)
C----- CALCULATE DIFFERENCES
                    CALL DIFF(ZA,ZM,ZU1)
                CALL DIFF(ZM, ZB,ZU2)
                CALL DIFF(ZC,ZM,ZL1)
                CALL DIFF(ZM,ZD,ZL2)
C---- IF TWO HALF DIFFERENCES ARE VERY DIFFERENT, IGNORE POINT
```

```
    ZT=ZU1/ZU2
    IF((ZT.LT.O.50).OR.(ZT.GT.1.50)) GOTO 800
    ZT=ZL1/ZL2
    IF((ZT.LT.0.50).OR.(ZT.GT.1.50)) GOTO 800
C----- COMPUTE AVERAGE DIFFERENCES
            ZU=(ZU1+ZU2)/2.0
            ZL=(ZL1+ZL2)/2.0
c----- check to SEE IF DIP IS NEGATIVE AND CORRECT IF SO
            IF(ZL.GT.ZPI) ZL=ZL-ZTWOPI
            ZMEM(VELSTR+K+J)=ZU/ZL
            ZDELTX=ABS(ZOFF(I+1)-ZOFF(I-1))/2.0
            ZMEM(VELSTR+K+J)=-1.0*ZMEM(VELSTR+K+J)*ZDELTX/ZDELTT
        continue
800
700
CONTINUE
c
c-.---
c
    DO 900 I=2,NEDMS-1,1
        K=(I-1)*RADIX
        COUNT=1
        DO 901 J=1,25
            ZPICKS(1, J,2)=0.0
            CONTINUE
    DO 1000 J=1,NAMPS-2,1
        IF(ABS(ZMEM(VELSTR+K+J)).LT.1.48) THEN
                    ZMEM(VELSTR+K+J)=0.0
                    GOTO 1000
                    ENDIF
            ZMEM(VELSTR+K+J)=ASIN(1.48/ZMEM(VELSTR+K+J))
            ZXSRC=(472-SPC)*25.0
            2TIME=J*0.004
            ZXRCV=ZXSRC+ZOFF(1)
            ZANGRC=ZMEM(VELSTR+K+J)
            IF(ZXRCV.GT.11780.0) GOTO }100
            ZPICKS(I, COUNT,1)=ZXSRC
            ZPICKS(1,COUNT,2)=ZTIME
            ZPICKS(I,COUNT,3)=FLOAT(CRP(I))
            ZPICKS(1, COUNT,4)=FLOAT(CRPTR(1))
            ZPICKS(1,COUNT,5)=ZXRCV
            ZPICKS(I,COUNT,6)=ZANGRC
            COUNT=COUNT+1
            continue
900 CONTINUE
C
C
    DO 1100 I=3,NEDMS-2
        DO 1110 J=1,25
            ZTIME=ZPICKS(I,J,2)
            IF(ZTIME.EQ.0.0) GOTO }110
            DO 1120 K=1,25
                IF(ABS(ZPICKS(I-1,K,2)-ZTIME).LT.5.000) THEN
                    KK=K
                    GOTO }113
                    ENDIF
                    CONTINUE
            GOTO }111
            DO 1140 K=1,25
                IF(ABS(ZPICKS(I+1,K,2)-ZTIME).LT.5.000) THEN
                    KKK=K
                    GOTO }115
                    ENDIF
                    CONTINUE
            GOTO }111
            ZK=ZPICKS(I,K,6)/((ZPICKS(I-1,KK,6)+ZPICKS(1+1,KKK,6))/2.0)
            WRITE(4,5000) ZPICKS(I,J,1),ZPICKS(I,J,2),
                                    INT(ZPICKS(I,J,3)),INT(ZPICKS(I,J,4)),
                                    ZPICKS(1,J,5),ZPICKS(I,J,6)
                continue
            CONTINUE
1 1 0 0
```



RETURN
$90100 \operatorname{WRITE}(6,90105)$
90105 FORMAT (' ',//,' Gото 99999

C
c
99999 STOP 15
END
SUBROUTINE DIFF(ZA,ZM,ZU1)
REAL ZA, ZM, ZU1, ZTWOPI
ZTWOPI $=8.0$ *ATAN ( 1.0 )
IF( ( $(Z A . G E \cdot 0.0)$.AND.(ZM.GE.0.0)).OR.
$+((Z A . L T .0 .0)$.AND. (ZM.LT. O.0) )) THEN
IF (ZA.GE.ZM) THEN
$\mathrm{ZU1}=\mathrm{ZA}-\mathrm{ZM}$
ELSE
ZU1=ZTWOPI - (ZM-ZA)
ENDIF
ENDIF
IF((ZA.GE.0.0).AND. (ZM.LT.0.0)) ZU1=ZA-ZM
IF((ZA.LT.0.0).AND.(ZM.GE.0.0)) ZU1=ZTWOPI-(ZM-ZA) RETURN
END

```
QPROCESS DC(GAUS),OPT(3), VECTOR(LEVEL(2),REPORT(XLIST)),
QPROCESS DIRECTIVE('*VD:')
```



```
c-..-- this subroutine traces a ray through a single cell
```



```
    subroutine raytrace ( \(x\), theta \(0, z, t\), triflag, tri, triloc)
c---- nq \(=\) maximum number of rows in velocity grid
c.-.-. mq = maximlm number of columns in velocity grid
c.-.- \(n=\) actual number of rows in velocity grid
c--.- m actual number of columns in velocity grid
c--.-- nint \(=\) spatial increment between grid rows
c.-.-. mint \(=\) spatial increment between grid columns
c.....
c---- vel(nq,mq) \(=\) values of velocity at grid points
c.-..- tol \(=1.0 \mathrm{~d}-10\)
c--.- pi, pion2,twopi \(=\) value of pi, pi divided by 2 and pi times 2
c---- i,,\(k=\) temporary integer variables
c--..- \(\mathrm{t} 1, \mathrm{t} 2, \mathrm{t} 3, \mathrm{t} 4, \mathrm{t} 5=\) temporary double-precision variables
c---- triflag \(=\).true. if triangle flat on top, .false. otherwise
c--.-- tri \((3,2)=\) grid points of current triangle
c--.-. - 1st dimension for the three corners
c--.-- - 2nd dimension for the column/row
c--.- triloc \((3,2)=\) grid point locations of current triangle (in metres)
c----- - 1st dimension for the three corners
c---- - 2nd dimension for the column/row
c--.-- vo \(=\) initial velocity in cell
c---- \(x, z=\) coordinates of the ray (in metres)
c----- t = cumulative travel time of the ray
c.-..- vx, vz= lateral and vertical components of the velocity gradient
c--.- for the current cell
c.-..- velang \(=\) angle the velocity gradient makes with the reference axis
c----- gradient= magnitude of the velocity gradient
c----- radius= radius of circular arc within current cell
c----- radius \(2=\) the square of the radius
c-...- straight \(=\).true. if ray is determined to be straight
c--.-- \(\mathrm{s}, \mathrm{e}=\) parameters of straight ray, \(\mathrm{z=s} . \mathrm{x}+\mathrm{e}\)
c-....
c--.--
c---- \(a, b=\) parameters of the local circular arc
c----- a2,b2= the squares of the above quantities
c--.- slope, \(c=\) parameters of sloping interface of triangle, \(z=s l o p e . x+c\)
c--..- slope2 \(=\) the square of slope
c---- intpoints= array of possible ray intersection points with cell boundary
c......
    - first entry is the \(x\) location, second is the \(z\) location
c-...- rslope, \(d=\) parameters of local reflector line segment, \(z=r\) lope. \(x+d\)
c----- rslope2 \(=\) the square of rslope
c---- clockwise= 1.0 if ray travelling clockwise on arc, -1.0 otherwise
c----- outangle \(=\) array holding exit angles at the intersection points
c--.- pathlength \(=\) array holding path lengths of arcs to exit points
c.-.-- exitpoint \(=\) flagged index of the exit point from cell
c.....
    real*4 nint,mint
    real* 4 vel
    real*8 pi,pion2,twopi
    real*8 tol
    real*8 \(x, z, t\)
    real*8 triloc
    real* \(8 \mathrm{vO}, \mathrm{vx}, \mathrm{vz}\)
    real*8 theta0, velang, gradient
    real*8 radius, radius2
    real*8 a,b,a2,b2
    real*8 slope, c,slope2
    real*8 s,e
    real*8 intpoints
    real*8 rslope, d
    real*8 t1, t2, t3, t4, t5
    real*8 val1, val2
    real*8 clockwise
    real*8 outangle, pathlength
```

```
    logical triflag,straight
    integer*4 loop
    integer*4 i,j,k
    integer*/4 n,m,nq,mq,nmax,mmax,max,ndat,dne
    integer*4 tri
    integer*4 exitpoint
    parameter ( }\textrm{nq}=90,\textrm{mq}=210\mathrm{ )
    dimension vel(nq,mq)
    dimension tri(3,2), triloc(3,2)
    dimension intpoints(8,2)
    dimension outangle(8),pathlength(8)
    common /velocity/vel
    common/constant/tol,pi,pion2, twopi
    common /conparm/n,m,nint,mint,nmax,mmax,max,ndat,dne
c
c---- set constants and initial variables
c
    t1=0.0d0
    t2=0.0d0
    t3=0.0d0
    t4=0.0d0
    t5=0.0d0
    straight=.false.
c
c---- check to see if ray is passing through a grid point and if so,
c----- shift it by one tenth of a millimetre
c
    if((dmod(x,dble(mint)).eq.0.0d0).and.
    + (dmod(z,dble(nint)).eq.0.0d0)) then
        x=x+.0001d0
        endif
c
c---- compute x and z components of the velocity gradient
c
    vx=(dble(vel(tri(2,2),tri(2,1))-vel(tri(1,2),tri(1,1))))
    + /dble(mint)
        if(triflag) then
        vz=(dble(vel(tri(3,2),tri(3,1))-vel(tri(2,2),tri(2,1))))
            /dble(nint)
        else
        vz=(dble(vel(tri(1,2),tri(1,1))-vel(tri(3,2),tri(3,1))))
            /dble(nint)
            endif
c
c----- compute initial velocity in cell
c
    v0=dble(vel(tri(1,2),\operatorname{tri}(1,1)))+(x-dble(tri(1,1)-1)*dble(mint))*vx
    +
                                    +(z-dble(tri(1,2)-1)*dble(nint))*vz
c
c---- compute angle velocity gradient makes with reference axes and
c----- the magnitude of the velocity gradient
c
    if(dabs(vz).lt.1.0d-10) then
        velang=dsign(pion2,vx)
    elseif(dabs(vx).lt.1.0d-10) then
    velang=pion2-dsign(pion2,vz)
    elseif(vz.gt.0.0d0) then
        velang=datan(vx/vz)
    else
        velang=dsign(1.0d0,vx)*(pi-datan(dabs(vx/vz)))
        endif
2 gradient=dsqrt(vz**2+vx**2)
c
c----- if the gradient is zero then flag the ray for straight-ray tracing
    if(gradient.lt.tol) then
    straight=.true.
    goto }1
    endif
```

c

```
c----- compute radius of ray in cell
c
    if(dabs(theta0-velang).gt.tol) then
        radius=v0/dabs(gradient*dsin(theta0-velang))
    else
        radius=1.0d10
        endif
3 radius2=radius*radius
c----- if the radius of the ray is very large, flag it as straight
c---- currently, larger than 1.0d6 means double precision cannot perform
    the required calulations for curved rays
c
    if(radius.lt.1.0d6) goto 12
    straight=.true.
c
c----- jump to here if ray flagged straight
c----- compute straight ray parameters, z=s.x+e
11 if(theta0.eq.0.0d0) goto 12
    s=1.0d0/dtan(theta0)
    e=z-s*x
12 continue
c
c---- here only if ray is curved
c----- compute parameters of the circle (x-a)**2+(z-b)**2=rad**2
c
c--.--.-- convert theta0 from range -pi to pi to range -pion2 to pion2
c---.---- this removes sense of direction of ray
    t2=theta0
    if(theta0.gt.pion2) t2=theta0-pi
    if(theta0. (t. -1.0d0*pion2) t2=theta0+pi
c---------- need to be careful of location of centre of circle
    val1=dcos(t2)
    val2=dsin(t2)
c
c---- check first of the two possible ray circle centres. Is vel zero?
c
    t1=1.0d0
    a=x+t1*val1*radius
    b=z-t1*val2*radius
    t3=(a-x)*vx+(b-z)*vz+v0
    if(dabs(t3).lt.1.0d0) goto 4
c
c----- check other possibility if the first was no good
c
    t1=-1.0d0
    a=x+t1*val1*radius
    b=z-t1*val2*radius
    t3=(a-x)*vx+(b-z)*vz+v0
    if(dabs(t3).ge.1.Od0) stop 'no suitable centre for ray'
    continue on
4 a2=a*a
    b2=b*b
13 continue
c---- find the exit point of the ray
c
c----- check the horizontal side of triangle for intersections
c
    if(straight) then
        if(theta0.eq.0.0d0) then
            t3=x
            t4=-1.0d0
            else
                t3=(triloc(1,2)-e)/s
                t4=-1.0d0
                endif
            goto }1
            endif
```

if (straight) then
if (theta0.eq. O.Od0) then
$\mathrm{t} 3=-1.0 \mathrm{~d} 0$
$t 4=-1.0 \mathrm{~d} 0$
goto 15
endif
t3 $=$ ** triloc $(3,1)+$ e
$t 4=-1.0 \mathrm{~d} 0$
goto 15
endif
if(triflag) then
t1=radius2-(triloc(2,1)-a)**2
else
t1=radius2-(triloc(1,1)-a)**2
endif
if(t1) $60,40,40$
t2=dsqrt(t1)
$\mathrm{t} 3=\mathrm{b}+\mathrm{t} 2$
$t 4=b-t 2$
if(.not.(triflag)) goto 50
if( (t3.lt.triloc $(2,2))$.or.(t3.gt.triloc $(3,2)))$ then
intpoints $(3,1)=-1.0 \mathrm{~d} 0$
intpoints $(3,2)=-1.0 \mathrm{~d} 0$
else
intpoints $(3,1)=$ triloc $(2,1)$
intpoints $(3,2)=t 3$
endif
if((t4.1t.triloc $(2,2))$.or.(t4.gt.triloc $(3,2)))$ then
intpoints $(4,1)=-1.0 \mathrm{dO}$
intpoints $(4,2)=-1.0 \mathrm{~d} 0$
else
intpoints $(4,1)=\operatorname{triloc}(2,1)$
intpoints $(4,2)=t 4$
endif
goto 70
t2=dsqrt(t1)
$t 3=a+t 2$
$t 4=a-t 2$
if((t3.lt.triloc(1, 1)).or.(t3.gt.triloc( 2,1$))$ ) then intpoints(1,1)=-1.0d0
intpoints (1,2)=-1.0d0
else
intpoints $(1,1)=t 3$
intpoints (1,2)=triloc(1,2)
endif
if((t4.lt.triloc(1, 1)).or.(t4.gt.triloc( 2,1$)$ )) then
intpoints $(2,1)=-1.0 \mathrm{~d} 0$
intpoints $(2,2)=-1.0 \mathrm{dO}$
else
intpoints $(2,1)=t 4$
intpoints $(2,2)=$ triloc $(1,2)$
endif
goto 30
intpoints (1,1)=-1.0d0
intpoints $(1,2)=-1.0 \mathrm{~d} 0$
intpoints $(2,1)=-1.0 \mathrm{~d} 0$
intpoints $(2,2)=-1.0 \mathrm{dO}$
continue
c---- check the vertical side of triangle for intersections
$\mathrm{t} 3=-1.0 \mathrm{~d} 0$
$t 4=-1.0 \mathrm{~d} 0$
goto 15
endif
iloc $(3,1)+e$
$4=-1.0 \mathrm{~d} 0$
endif
if(triflag) then
t1=radius2-(triloc(2,1)-a)**2
else
t1=radius2-(triloc(1,1)-a)**2
endif
if(t1) $60,40,40$
t2=dsqrt(t1)
$\mathrm{t} 3=\mathrm{b}+\mathrm{t} 2$

- 2
if((t3.lt.triloc $(2,2))$.or.(t3.gt.triloc $(3,2)))$ then intpoints $(3,1)=-1.0 \mathrm{~d} 0$
else
intpoints $(3,1)=$ triloc $(2,1)$
endif
if((t4.lt.triloc(2,2)).or.(t4.gt.triloc(3,2))) then intpoints $(4,1)=-1.0 \mathrm{dO}$
intpoints $(4,2)=-1.0 \mathrm{~d} 0$
intpoints $(4,1)=\operatorname{triloc}(2,1)$
endif
goto 70
if((t3.gt.triloc $(2,2))$.or. (t3.lt.triloc $(3,2)))$ then
t1=radius2-(triloc(1,2)-b)**2
if(t1) 20,10,10
intpoints $(3,1)=-1.0 \mathrm{~d} 0$
intpoints $(3,1)=-1.0 \mathrm{~d} 0$
else
intpoints $(3,1)=$ triloc (1, 1)
intpoints $(3,2)=t 3$
endif
if((t4.gt.triloc $(2,2))$.or. (t4.lt.triloc $(3,2)))$ then intpoints $(4,1)=-1.0 \mathrm{~d} 0$
intpoints $(4,2)=-1.0 \mathrm{~d} 0$
else
intpoints $(4,1)=\operatorname{tri} \operatorname{loc}(1,1)$
intpoints $(4,2)=t 4$
endif
goto 70
intpoints $(3,1)=-1.0 \mathrm{~d} 0$
intpoints $(3,2)=-1.0 \mathrm{dO}$
intpoints $(4,1)=-1.0 \mathrm{~d} 0$
intpoints $(4,2)=-1.0 \mathrm{~d} 0$
continue
c
c---.
check the sloping side of the triangle. First find the parameters of the of the line $z=$ slope $. x+c$, then solve quadratic similarly to above.
if(triflag) then
slope=(triloc(1,2)-triloc(3,2))/(triloc(1,1)-triloc(3,1))
else
slope $=(\operatorname{triloc}(2,2)$-triloc $(3,2)) /(\operatorname{triloc}(2,1)$-triloc $(3,1))$
endif
c=triloc (3,2)-slope*triloc $(3,1)$
if(straight) then
if(theta0.eq. 0.0d0) then
t3 $=\mathrm{x}$
$t 4=-1.0 \mathrm{~d} 0$
goto 16
endif
t3 $=(\mathrm{c}-\mathrm{e}) /(\mathrm{s}$-slope)
$t 4=-1.0 \mathrm{~d} 0$
goto 16
endif
slope2=slope**2
t $\left.1=(\text { slope* }(c-b)-a)^{* * 2-(s l o p e 2+1.0 d 0) *(a 2+(c-b) * * 2-r a d i u s ~} 2\right)$
if(t1) $90,80,80$
t2=dsqrt(t1)
$t 3=((a-$ slope* $(c-b))+t 2) /($ slope $2+1.0 \mathrm{~d} 0)$
t4 = ((a-slope* $(c-b))-$ t2) $/($ slope2+1.0d0)
16 if((t3.lt.triloc(1,1)).or.(t3.gt.triloc(2,1))) then
intpoints $(5,1)=-1.0 \mathrm{~d} 0$
intpoints $(5,2)=-1.0 \mathrm{~d} 0$
else
intpoints $(5,1)=t 3$
intpoints $(5,2)=$ slope ${ }^{*} t 3+c$
endif
if((t4.lt.triloc(1, 1)).or.(t4.gt.triloc(2,1))) then
intpoints $(6,1)=-1.0 \mathrm{~d} 0$
intpoints $(6,2)=-1.0 \mathrm{~d} 0$
else
intpoints $(6,1)=t 4$
intpoints $(6,2)=$ slope* $t 4+$ c
endif
goto 100
90
intpoints $(5,1)=-1.0 \mathrm{~d} 0$
intpoints $(5,2)=-1.0 \mathrm{dO}$
intpoints $(6,1)=-1.0 \mathrm{~d} 0$
intpoints $(6,2)=-1.0 \mathrm{~d} 0$
continue
c
c---- compute exit angles for all the possible intersection points
c---- need to determine sense of rotation of the ray (clockwise or not)
c--.- flag anticlockwise rays with a negative sign
if(straight) goto 18
if(theta0.eq.0.0d0) then
if(a.lt.x) then
clockwise=1.0d0
else
clockwise=-1.0d0
endif
goto 145
endif
if(dabs(theta0).eq.pi) then


## if(a.lt.x) then

 clockwise=-1.0d0else
clockwise=1.0d0
endif
goto 145
endif
if (dabs(theta0).eq.pion2) then
if(b.lt.z) then
clockwise=dsign(1.0d0,-1.0d0* theta0)
else
clockwise=dsign(1.0d0, theta0)
endif
goto 145
endif
if((a.ge.x).and.(b.ge.z)) then
clockwise=dsign(1.0d0, theta0)
goto 145
endif
if((a.ge. $x$ ).and.(b.lt.z)) then
clockwise=dsign(1.0d0,-1.0d0* theta0)
goto 145
endif
if((a.lt. $x$ ). and. (b.ge.z)) then
clockwise=dsign(1.0d0, theta0)
goto 145
endif
if ( $(a . l t, x)$.and. (b.lt. $z)$ ) then
clockwise=dsign(1.0d0,-1.0d0*theta0)
endif
do $140 \quad i=1,6$
if(intpoints(i,1).eq. -1.0 d 0 ) goto 140
if(a-intpoints(i, 1)) $170,160,150$
outangle(i)=dasin((intpoints(i,2)-b)/radius)
goto 146
outangle(i)=pion2
goto 146
outangle(i)=dasin((b-intpoints(i,2))/radius)
if((a.gt. intpoints $(i, 1))$.and. (b.gt.intpoints $(i, 2))$ ) then
if(clockwise.gt. O.Od0) outangle(i)=outangle(i)+pi
elseif((a.gt.intpoints(i,1)), and. (b. (t.intpoints( $i, 2))$ ) then if(clockwise.gt.0.0d0) outangle(i)=outangle(i)-pi
elseif( (a.lt.intpoints $(i, 1))$.and. (b.gt.intpoints $(i, 2)))$ then
if(clockwise.lt.0.0d0) outangle(i)=outangle(i)-pi
elseif((a.lt.intpoints $(i, 1))$. and. (b.lt.intpoints $(i, 2)))$ then
if(clockwise.lt.0.0d0) outangle(i)=outangle(i)+pi
elseif(a.eq.intpoints(i,1)) then
outangle( $i$ )=dsign(pion2, -1 .0d0*clockwise)
elseif(b.eq.intpoints (i,2)) then
if(clockwise.lt.0.0d0) outangle(i)=outangle(i)-pi
endif
if (outangle(i).lt.0.0d0) outangle(i)=twopi+outangle(i)
outangle(i)=clockwise*outangle(i)
continue
continue
c
c--- compute path lengths to each of the intersection points
c---- remembering to compute length in direction of travel
c---- also find nearest intpoint and set its pathlength to 1.0 d 20
c
$\mathrm{t} \boldsymbol{1}=$ theta0
$\mathrm{t} 3=1.0 \mathrm{~d} 70$
$\mathrm{k}=0$
if(theta0.lt.0.0d0) t1=twopi+theta0
do $180 \quad i=1,6$
if(intpoints( $i, 1$ ).eq. -1.0 d 0 ) then
pathlength ( $i$ ) $=1.0 \mathrm{~d} 20$
elseif(straight) then
pathlength $(i)=\operatorname{dsqrt}\left((\text { intpoints }(i, 1)-x)^{\text {** }} 2^{+}\right.$
(intpoints $(i, 2)-z$ )**2)
else
if (outangle(i).lt. O.0d0) then

```
                    t2=dabs(outangle(i))-t1
                else
                    t2=t1-outangle(i)
                    endif
                if(t2.lt.0.OdO) t2=twopi+t2
                pathlength(i)=radius*t2
                endif
        t4=(x-intpoints(i,1))**2+(z-intpoints(i,2))**2
        if(t4.lt.t3) then
            t3=t4
            k=i
            endif
        continue
    pathlength(k)=1.0d20
c
c---- choose shortest pathlength left and flag that exit point
c
    exitpoint=0
    t1=1.0d20
    do }195\textrm{i}=6,1,-
        if(pathlength(i).lt.t1) then
                            t1=pathlength(i)
                    exitpoint=i
                    endif
195 continue
C
c----- compute time through this cell and add to ray's total time
    if(straight) then
            t1=((intpoints(exitpoint,1)-x)*vx)+
            ((intpoints(exitpoint,2)-z)*vz)+v0
c----- nb: harmonic mean
            t1=(1/t1+1/v0)/2
            t1=pathlength(exitpoint)*t1
            t=t+t1
        else
            t2=dabs(outangle(exitpoint))
            if(t2.gt.pi) t2=t2-twopi
            t4=dabs(velang-t2)
            t5=dabs(velang-theta0)
            if(t4.gt.pi) t4=dabs(t4-twopi)
            if(t5.gt.pi) t5=dabs(t5-twopi)
            t1=(1/dabs(gradient))*dlog(dtan(t4/2.0d0)/dtan(t5/2.0d0))
            t=t+t1
            endif
c
c-\cdots-- reset necessary parameters for completion of this triangle
c
    x=intpoints(exitpoint,1)
    z=intpoints(exitpoint,2)
    if(.not.(straight)) then
                    theta0=dabs(outang(e(exitpoint))
                    if(theta0.gt.pi) theta0=theta0-twopi
                    endif
        straight=.false.
c
c---- check to see if ray is passing through a grid point and if so,
c--.-- shift it by one tenth of a millimetre and compute new cell
c
    if((dmod(x,dble(mint)).lt.tol).and.
            (dmod(z,dble(nint)).(t.tol)) then
            if(triflag) then
                    if(c(x-triloc(1,1)).lt.tol).and.((z-triloc(1,2)).lt.tol)) then
                    x=triloc(1,1)
                    z=triloc(1,2)+0.0001d0
                    tri(1,1)=tri(3,1)-1
                    tri(1,2)=tri(3,2)
                    tri}(2,1)=tri(1,1)+
                    tri(2,2)=tri(1,2)
                    tri(3,1)=tri(1,1)
                    tri(3,2)=tri(1,2)-1
                triloc(1,1)=triloc(3,1)-dble(mint)
```

```
    triloc(1,2)=triloc(3,2)
    triloc(2,1)=triloc(1,1)+dble(mint)
    triloc(2,2)=triloc(1,2)
    triloc(3,1)=triloc(1,1)
    triloc(3,2)=triloc(1,2)-dble(nint)
elseif(((x-triloc(2,1)).lt.tol).and.((z-triloc(2,2)).lt.tol)) then
    x=tritoc(2,1)
    z=triloc(2,2)+0.0001d0
    tri(1,1)=tri}(3,1
    tri(1,2)=tri(3,2)
    tri(2,1)=tri(1,1)+1
    tri(2,2)=tri(1,2)
    tri(3,1)=tri(1,1)
    tri(3,2)=tri(1,2)-1
    triloc(1,1)=triloc(3,1)
    triloc(1,2)=triloc(3,2)
    triloc(2,1)=triloc(1,1)+dble(mint)
    triloc(2,2)=triloc(1,2)
    triloc(3,1)=triloc(1,1)
    triloc(3,2)=triloc(1,2)-dble(nint)
elseif(((x-triloc(3,1)).lt.tol).and.((z-triloc(3,2)).lt.tol)) then
    x=triloc(3,1)
    z=triloc(3,2)-0.0001d0
    tri(1,1)=tri(3,1)
    tri(1,2)=tri(3,2)
    tri(2,1)=tri(1,1)+1
    tri(2,2)=tri(1,2)
    tri(3,1)=tri(1,1)
    tri(3,2)=tri(1,2)-1
    triloc(1,1)=triloc(3,1)
    triloc(1,2)=triloc(3,2)
    triloc(2,1)=triloc(1,1)+dble(mint)
    triloc(2,2)=triloc(1,2)
    triloc(3,1)=triloc(1,1)
    triloc(3,2)=triloc(1,2)-dble(nint)
    endif
else
    if(((x-triloc(1,1)).lt.tol).and.((z-triloc(1,2)).lt.tol)) then
    x=triloc(1,1)+0.0001d0
    z=triloc(1,2)
    tri(3,1)=tri(2,1)
    tri(3,2)=tri(2,2)+1
    triloc(3,1)=triloc(2,1)
    triloc(3,2)=triloc(2,2)+dble(nint)
elseif(((x-triloc(2,1)).lt.tol).and.((z-triloc(2,2)).lt.tol)) then
    x=triloc(2,1)-0.0001d0
    z=triloc(2,2)
    tri(3,1)=tri(2,1)
    tri(3,2)=tri(2,2)+1
    triloc(3,1)=triloc(2,1)
    triloc(3,2)=triloc(2,2)+dble(nint)
elseif(((x-triloc(3,1)).lt.tol).and.((z-triloc(3,2)).lt.tol)) then
    x=triloc(3,1)
    z=triloc(3,2)+0.0001d0
    tri(1,1)=tri(3,1)-1
    tri(1,2)=tri(3,2)
    tri(2,1)=tri(1,1)+1
    tri(2,2)=tri(1,2)
    tri(3,1)=tri(2,1)
    tri(3,2)=tri(2,2)+1
    triloc(1,1)=triloc(3,1)-dble(mint)
    triloc(1,2)=triloc(3,2)
    triloc(2,1)=triloc(1,1)+dble(mint)
    triloc(2,2)=triloc(1,2)
    triloc(3,1)=triloc(2,1)
    triloc(3,2)=triloc(2,2)+dble(nint)
    endif
    endif
triflag=.not.(triflag)
goto 1000
endif
if((exitpoint.eq.1).or.(exitpoint.eq.2)) then
```

```
    if(triflag) then
    tri(3,1)=tri(3,1)-1
    tri(3,2)=tri(3,2)-2
    triloc(3,1)=triloc(3,1)-dble(mint)
    triloc(3,2)=triloc(3,2)-2.0d0*dble(nint)
else
    tri(3,1)=tri(3,1)+1
    tri(3,2)=tri(3,2)+2
    triloc(3,1)=triloc(3,1)+dble(mint)
    triloc(3,2)=triloc(3,2)+2.0d0*dble(nint)
    endif
triflag=.not.(triflag)
endif
if((exitpoint.eq.3).or.(exitpoint.eq.4)) then
    if(triflag) then
    tri(1,1)=tri(3,1)
    tri(1,2)=tri(3,2)
    tri(2,1)=tri(1,1)+1
    tri(2,2)=tri(1,2)
    tri}(3,1)=tri(1,1
    tri(3,2)=tri(1,2)-1
    triloc(1,1)=triloc(3,1)
    triloc(1,2)=triloc(3,2)
    triloc(2,1)=triloc(1,1)+dble(mint)
    triloc(2,2)=triloc(1,2)
    triloc(3,1)=triloc(1,1)
    triloc(3,2)=triloc(1,2)-dble(nint)
    else
        tri(1,1)=tri(3,1)-1
        tri(1,2)=tri(3,2)
        tri(2,1)=tri(1,1)+1
        tri(2,2)=tri(1,2)
        tri(3,1)=tri(1,1)+1
        tri(3,2)=tri(1,2)+1
        triloc(1,1)=triloc(3,1)-dble(mint)
        triloc(1,2)=triloc(3,2)
        triloc(2,1)=triloc(1,1)+dble(mint)
        triloc(2,2)=triloc(1,2)
        triloc(3,1)=triloc(1,1)+dble(mint)
        triloc(3,2)=triloc(1,2)+dble(nint)
        endif
    triflag=.not.(triflag)
    endif
if((exitpoint.eq.5).or.(exitpoint.eq.6)) then
if(triflag) then
    tri(1,1)=tri(3,1)-1
    tri(1,2)=tri(3,2)
    tri(2,1)=tri(1,1)+1
    tri(2,2)=tri(1,2)
    tri(3,1)=tri(1,1)
    tri(3,2)=tri(1,2)-1
    triloc(1,1)=triloc(3,1)-dble(mint)
    triloc(1,2)=triloc(3,2)
    triloc(2,1)=triloc(1,1)+dble(mint)
    triloc(2,2)=triloc(1,2)
    triloc(3,1)=triloc(1,1)
    triloc(3,2)=triloc(1, 2)-dble(nint)
else
    tri(1,1)=tri(3,1)
    tri(1,2)=tri(3,2)
    tri(2,1)=tri(1,1)+1
    tri(2,2)=tri(1,2)
    tri(3,1)=tri(1,1)+1
    tri}(3,2)=\operatorname{tri}(1,2)+
    triloc(1,1)=triloc(3,1)
    triloc(1,2)=triloc(3,2)
    triloc(2,1)=triloc(1, 1)+dble(mint)
    triloc(2,2)=triloc(1,2)
    triloc(3,1)=triloc(1, 1)+dble(mint)
    triloc(3,2)=triloc(1,2)+dble(nint)
    endif
triflag=.not.(triflag)
```


## endif

1000 return end
@PROCESSqDC(GAUS), OPT(3),VECTOR(LEVEL(2), REPORT(XLIST)), @PROCESS DIRECTIVE('*VD:')
c--.-- subroutine to compute the frechet derivatives for submission as
c----- the response matrix to MEM NL
c----- this subroutine uses Semi-Analytic Derivative Estimate no 1
c----- uses Chuck Sword's XERR statistic
c---- this routine is called for each ray involved in the data times

c---- index $=$ index number of ray as stored in geom file
c--.-- sloc1 = surface location of ray1
c--.-- tang1 $=$ take off angle of ray1
c--.-- sloc2 $=$ surface location of ray2
c---- tang2 $=$ take off angle of ray2
c----- datatime $=$ actual measured time of ray
c----- xerr $=$ Chuck Sword's statistic
c--...
c----- nq $=$ maximum number of rows in velocity grid
c----- mq = maximum number of columns in velocity grid
c---- $n=$ actual number of rows in velocity grid
c----- m = actual number of columns in velocity grid
c----- nint = spatial increment between grid rows
c----- mint $=$ spatial increment between grid columns
c--...
c----- nmaxq $=$ maximum number of velocity nodes
c----- maxq $=$ maximum number of data rays
c----- $\operatorname{maxq}=$ maximum of $n \operatorname{maxq}$ and mmaxq
c----- nmax $=$ actual number of velocity nodes
c----- max $=$ actual number of data rays
c----- max $=$ maximum of $n_{m a x}$ and max
c----
c---- vel $(n q, m q)=$ values of velocity at grid points
c----- ray1travel $\left(n q^{*} m q, 6\right)=$ array that stores cell,theta0, $x, z, t$ for ray1's path
c----- num $\quad=$ number of points in ray1travel
c----- ray2travel $\left(n q^{*} m q, 6\right)=$ array that stores cell, theta0, $x, z, t$ for ray2's path
c---- num2 $=$ number of points in ray2travel
c.-..-
c----- i, j,k= temporary integer variables
c----- t1, t2, t3, t4, t5= temporary double-precision variables
c----
c---- tri $=$ coordinate indices of the apex of the current triangle
c----
c---- row $=$ the row of the current cell
c----- col $=$ the column of the current cell
c---- trinode1,2,3 = velocity node indices of apices of current triangle c----- rec1 = integer variable for record numbers
c-----
c---- $R(n q * m q)=$ temporary frechet derivative array
c--...
c--------......................
real*4 nint,mint
real*4 ray1travel, ray2travel
real*4 sloc1,tang1, sloc2,tang2
real*4 vel
real*4 datatime, xerr, newxerr
real*4 R
real*8 triloc, tmptriloc
real*8 thetaV, lenV
real*8 xout, zout, thetaout, timeout
real*8 finalx1,finalx2,finalz,dtdz1,dxdz1,dtdz2,dxdz2
real*8 finalv1, finalv2
real*8 newthV, newx1,newx2,newz,newdxdz1,newdxdz2
real*8 newdtdz1,newdtdz2
real* 8 t1, t2, t3, t4, t5
logical triflag, tmptriflag
integer*4 i, j,k,ndat, ndatq
integer*4n,m,dne,nq,mq
integer*4 index

```
integer*4 nmax,mmax,max,nmaxq, mmaxq, maxq
integer*4 num,num2
integer*4 row,col
integer*4 tri,tmptri
integer*4 trinode
integer*4 rec1
integer*4 p,mult,rem,block,nc
parameter( }\textrm{nq}=90,\textrm{mq}=210
parameter(nmaxq=17000,mmaxq=4200, maxq=17000)
parameter(ndatq=4200)
    dimension vel(nq,mq)
    dimension ray1travel(nq*mq,6),ray2travel(nq*mq,6)
    dimension R(nq*mq)
    dimension tri(3,2),triloc(3,2)
    dimension tmptri(3,2),tmptriloc(3,2)
    dimension trinode(3)
    common /temp/nc
    common /velocity/vel
    common /travel/ray1travel,num, ray2travel,num2
    common /sade/finalx1,finalx2, finalz,dtdz1,dxdz1,dtdz2,dxdz2,
+
    finalv1,finalv2
    common /conparm/n,m,nint,mint,nmax,mmax,max, ndat, dne
c----- set constants and initial variables
    t1=0.0d0
    t2=0.0d0
    t3=0.0d0
    t4=0.0d0
    t5=0.0d0
c----- compute all velocity nodes affected by the rays
c----- begin working along path of ray1
```

c
c
c
c
if(num.lt.2) goto 22222
do $1000 \mathrm{i}=1$, num-1
if(ray1travel( $i+1,4)$.ge.finalz) goto 1000
c
c---- first compute the 'tri' nodes affecting this cell
c
$\mathrm{p}=$ int(ray1travel(i,1))
rem $=\bmod \left(p,\left(2^{*}(m-1)\right)\right)$
mul $t=(p-r e m) /\left(2^{*}(m-1)\right)$
row=mult+1
if(rem.eq.0) row=row-1
if(rem.gt.0) then
col=int $(($ rem-1)/2.0) +1
else
col=m-1
endif
if(mod(ray1travel $(i, 1), 2.0 e 0) . e q \cdot 0.0 e 0)$ then
$\operatorname{tri}(1,1)=\mathrm{col}$
$\operatorname{tri}(1,2)=$ row
$\operatorname{tri}(2,1)=\mathrm{col}+1$
$\operatorname{tri}(2,2)=$ row
$\operatorname{tri}(3,1)=\operatorname{col}+1$
tri $(3,2)=r o w+1$
else
tri $(1,1)=\mathrm{col}$
tri(1,2)=row+1
$\operatorname{tri}(2,1)=c o l+1$
tri $(2,2)=$ row +1
$\operatorname{tri}(3,1)=\operatorname{col}$
tri $(3,2)=$ row
endif
triloc(1, 1)=dble((tri(1,1)-1)*mint)
triloc(1,2)=dble((tri(1,2)-1)*nint)
triloc $(2,1)=d b l e((\operatorname{tri}(2,1)-1) * \operatorname{mint})$
triloc $(2,2)=\mathrm{dble}((\operatorname{tri}(2,2)-1)$ *nint $)$

```
triloc(3,1)=dble((tri(3,1)-1)*mint)
triloc(3,2)=dble((tri(3,2)-1)*nint)
if(tri(1,2).lt.tri(3,2)) then
    triflag=.true.
else
    triflag=.false.
    endif
```

c
c---- compute the corresponding velocity node indices
c
trinode(1) $=($ tri $(1,2)-1) * m+\operatorname{tri}(1,1)$
trinode $(2)=(\operatorname{tri}(2,2)-1) * m+\operatorname{tri}(2,1)$
trinode $(3)=(\operatorname{tri}(3,2)-1) * m+\operatorname{tri}(3,1)$
c
c---- compute derivatives for these nodes and add to R matrix
c
c---- first re-trace through cell with perturbed velocity to find
c----- the effect on exitangle and time
c---- do this for each of the 3 trinodes
c
c--- compute parameters of the vector $V$
t1 $=\mathrm{dble}$ (ray1travel ( $\mathrm{i}+1,3$ ))
$t 3=d b l e($ ray1travel $(i+1,4)$ )
thetaV $=\operatorname{datan}(($ final $x 1-t 1) /($ finalz $-t 3))$
lenV=dsqrt((finalx1-t1)**2+(finalz-t3)**2)
do $1010 \mathrm{k}=1$, 3
tmptri(1,1)=tri(1,1)
tmptri(1,2)=tri(1,2)
tmptri $(2,1)=\operatorname{tri}(2,1)$
tmptri(2,2)=tri(2,2)
tmptri $(3,1)=\operatorname{tri}(3,1)$
tmptri(3,2)=tri(3,2)
tmptriloc(1, 1)=triloc(1, 1)
tmptriloc (1,2)=triloc (1,2)
tmptriloc $(2,1)=$ triloc $(2,1)$
tmptriloc $(2,2)=$ triloc $(2,2)$
tmptriloc $(3,1)=$ triloc $(3,1)$
tmptriloc $(3,2)=$ triloc $(3,2)$
tmptriflag=triflag
t1=dble(ray1travel(i,3))
t2=dble(ray1travel(i,2))
t $3=$ dble(ray1travel ( 1,4 ))
$\mathrm{t} 4=\mathrm{dble}($ ray1travel $(\mathrm{i}, 5))$
t5 =0.0d0
$\operatorname{vel}(\operatorname{tri}(k, 2), \operatorname{tri}(k, 1))=\operatorname{vel}(\operatorname{tri}(k, 2), \operatorname{tri}(k, 1)) * 1.0025$
call raytrace(t1, t2, t3, t4, tmptriflag, tmptri, tmptriloc)
$\operatorname{vel}(\operatorname{tri}(k, 2), \operatorname{tri}(k, 1))=\operatorname{vel}(\operatorname{tri}(k, 2), \operatorname{tri}(k, 1)) / 1.0025$
xout=t1
zout= t 3
thetaout=t2
timeout=t4
c----- adjust thetaV and compute new end point
$t 1=$ (thetaout $-\mathrm{dble}($ ray $/$ travel $(\mathrm{i}+1,2))$ )
newthV=thetaV+t1
newx $1=x$ out + lenV*dsin(newthV)
newz=zout+lenV*dcos(newthV)
c--.- compute new deriv's and allow for time/depth correction and
c---.- then compute partial frechet deriv for this node
newdxdz1=dtan(datan(dxdz1)+t1)
if(dabs(1/(finalv1*dtdz1)).gt.1.0d0) then
newdtdz1=1/(finalv1*dcos(t1))
else
newdtdz1=1/(finalv1*dcos(dacos(1/(finalv1*dtdz1))+t1))
endif
t5 = dsqrt $((\text { xout }-\mathrm{dble} \text { (ray1travel }(\mathrm{i}+1,3)))^{* * 2+(\text { zout }}$
dble(ray1travel $\left.(i+1,4)))^{* *} 2\right)$
if(t5.gt.1.0d0) then
newz=newz+(timeout-ray1travel( $\mathbf{i}+1,5$ ))/newdtdz1
endif
$t 2=(\text { newz }- \text { final } z)^{*} d t d z 2+$ (timeout-ray1travel $(i+1,5)$ )
newx2=(newz-finalz)*dxdz2+finalx2

```
            t3=-1.0d0*t2/(newdtdz1+dtdz2)
                    t4=((t3*dxdz2)+newx2)-((t3*newdxdz1)+newx1)
                    if(dabs(t4-dble(xerr)).(t.10.0d0)
    +
R(trinode(k))=R(trinode(k))+sngl(t4-dble(xerr))/
                            (vel(tri(k,2),tri(k,1))*0.0025)
                            if(dabs(t4-dble(xerr)).ge.10.0d0) nc=nc+1
                            continue
                            continue
1000
now do the same for ray2
c-
22222 if(num2.(t.2) goto 99999
do 2000 i=1,num2-1
                    if(ray2travel(i+1,4).ge.finalz) goto 2000
c
c----- first compute the 'tri' nodes affecting this cell
c
    p=int(ray2travel(i,1))
    rem=mod(p,(2*(m-1)))
    mult=(p-rem)/(2*(m-1))
    row=mult+1
    if(rem.eq.0) row=row-1
    if(rem.gt.0) then
    col=int((rem-1)/2.0)+1
else
        col=m-1
        endif
    if(mod(ray2travel(i,1),2.0e0).eq.0.0e0) then
        tri(1,1)=col
        tri(1,2)=row
        tri(2,1)=col+1
        tri(2,2)=row
        tri(3,1)=col+1
        tri(3,2)=row+1
        etse
            tri(1,1)=col
            tri(1,2)=row+1
            tri(2,1)=col+1
            tri(2,2)=row+1
            tri(3,1)=col
            tri(3,2)=row
            endif
            triloc(1,1)=dble((tri(1,1)-1)*mint)
            triloc(1,2)=dble((tri(1,2)-1)*nint)
            triloc(2,1)=dble((tri(2,1)-1)*mint)
            triloc(2,2)=dble((tri(2,2)-1)*nint)
            triloc(3,1)=dble((tri(3,1)-1)*mint)
            triloc(3,2)=dble((tri(3,2)-1)*nint)
            if(tri(1,2).lt.tri(3,2)) then
            triflag=.true.
                else
            triflag=.false.
            endif
c
c--.-- compute the corresponding velocity node indices
c
    trinode(1)=(tri(1,2)-1)*m+tri(1,1)
            trinode(2)=(tri (2,2)-1)*m+tri}(2,1
            trinode(3)=(tri (3,2)-1)*m+tri (3,1)
c
c----- compute derivatives for these nodes and add to R matrix
c
c---- first re-trace through cell with perturbed velocity to find
c----- the effect on exitangle and time
c----- do this for each of the 3 trinodes
c
c----- compute parameters of the vector V
    t1=dble(ray2travel(i+1,3))
    t3=dble(ray2travel( }\textrm{i}+1,4)\mathrm{ )
    thetaV=datan((final }\times2-t1)/(finalz-t3)
    lenV=dsqrt((finalx2-t1)**2+(finalz-t3)**2)
```

```
    do 2010 k=1,3
    tmptri(1,1)=tri(1,1)
    tmptri(1,2)=tri(1,2)
    tmptri(2,1)=tri(2,1)
    tmptri(2,2)=tri(2,2)
    tmptri(3,1)=tri}(3,1
    tmptri(3,2)=tri(3,2)
    tmptriloc(1,1)=triloc(1,1)
    tmptriloc(1,2)=triloc(1,2)
    tmptriloc(2,1)=triloc(2,1)
    tmptriloc(2,2)=triloc(2,2)
    tmptriloc(3,1)=triloc(3,1)
    tmptriloc(3,2)=triloc(3,2)
    tmptriflag=triflag
    t1=dble(ray2travel(i,3))
    t2=dble(ray2travel(i,2))
    t3=dble(ray2travel(i,4))
    t4=dble(ray2travel(i,5))
    t5=0.0d0
    vel(tri(k,2),tri(k,1))=vel(tri(k,2),tri(k,1))*1.0025
    call raytrace(t1, t2, t3, t4,tmptriflag,tmptri,tmptriloc)
    vel(tri(k,2),tri(k,1))=vel(tri(k,2),tri(k,1))/1.0025
    xout=t1
    zout=t3
    thetaout=t2
    timeout=t4
c---- adjust thetaV and compute new end point
            t1=(thetaout-dble(ray2travel(i+1,2)))
            newthV=thetaV+t1
            newx2=xout+lenV*dsin(newthV)
            newz=zout+lenV*dcos(newthV)
c----- compute new deriv's and allow for time/depth correction and
c---- then compute partial frechet deriv for this node
            newdxdz2=dtan(datan(dxdz2)+t1)
            if(dabs(1/(finalv2*dtdz2)).gt.1.0d0) then
                newdtdz2=1/(finalv2*dcos(t1))
            else
                newdtdz2=1/(finalv2*dcos(dacos(1/(finalv2*dtdz2))+t1))
                    endif
            t5=dsqrt((xout-dble(ray2travel(i+1,3)))**2+(zout-
                    dble(ray2travel(i+1,4)))**2)
            if(t5.gt.1.0d0) then
                    newz=newz+(timeout-ray2travel(i+1,5))/newdtdz2
                    endif
            t2=(newz-finalz)*dtdz1+(timeout-ray2travel(i+1,5))
            newx1=(newz-finalz)*dxdz1+finalx1
            t3=-1.0d0*t2/(newdtdz2+dtdz1)
            t4=((t3*newdxdz2)+newx2)-((t3*dxdz1)+newx1)
            if(dabs(t4-dble(xerr)).lt.10.0d0)
            R(trinode(k))=R(trinode(k))+sngl(t4-dble(xerr))/
                (vel(tri(k,2),tri(k,1))*0.0025)
            if(dabs(t4-dble(xerr)).ge.10.0d0) nc=nc+1
            continue
        continue
9 9 9 9 9 ~ r e t u r n ~
    end
```

@PROCESS DC(GAUS), OPT(3), VECTOR(LEVEL(2), REPORT(XLIST)), @PROCESS DIRECTIVE('*VD:')
c----- subroutine to compute the weights based on the Xerr's for each ray ..... c---- currently the Xerr threshold is 300 metres

c----- the preliminary comments in METOMOS6 describe the variables in this
c----- subroutine
c----
real*4 F, sig, ai ,bi, XXerr, XXerr2
real*4 pi, pi6, pi18, t1, ts, tc
real* 4 nint,mint
integer ${ }^{*} 4$ ndat, ndatq
integer* $4 i, j, k, n$, $m$, nmax, mmax, max , dne
parameter(ndatq=4200)
dimension $F$ (ndatq), sig(ndatq)
dimension ai (ndatq), bi (ndatq), XXerr(ndatq), XXerr2(ndatq)
common /var/sig, ai, bi,XXerr, XXerr2
common /conparm/n,m,nint,mint, nmax, mmax, max, ndat, dne
c
c---- set constants and initial variables
c
$\mathrm{pi}=4.0$ *atan ( 1.0 )
pi6=-1.0*pi/600.0
pi $18=-1.0^{*}\left(\mathrm{pi}^{* * 2}\right) / 180000.0$
c
c-----compute the sig(k) values using a modified Hanning window function
c-----also compute the ai's, bi's, and set Xerr and Xerr2 (see T.N.26)
c
do $100 \mathrm{k}=1$, ndat $t 1=p i * F(k) / 300.0$
$t s=\sin (t 1)$
$t c=\cos (t 1)$
ai $(k)=$ pi6*ts* (0.75+0.25*tc)
bi $(k)=$ pi 18* (0.75*tc+0.25*tc**2-0.25*ts**2)
$X X \operatorname{Xer}(k)=F(k)$
XXerr2(k) $=F(k) * * 2$
if( $F(k)$.ge.300.0) then
$\operatorname{sig}(k)=0.25$
ai $(k)=0.0$
$\operatorname{bi}(k)=0.0$
goto 100 endif
$\operatorname{sig}(k)=0.75+0.25 * t c$
$\operatorname{sig}(k)=\operatorname{sig}(k) * * 2$
continue
return
end
aprocess dC(GAUS), OPT(3), VECTOR(LEVEL(2),REPORT(XLIST)), @PROCESS DIRECTIVE('*VD:')

```
c---- subroutine to trace two rays through medium divided into triangles
c----- both traces have known starting locations and initial angles
c-\cdots-- traces both rays through depth levels until data time is exceded ....-
c---- so that Chuck Sword's 'xerr' statistic can be calculated
c----- all real calculations in double-precision
    subroutine xerrrt(sloc1,tang1,sloc2,tang2,datatime, xerr,ray)
c---- sloc1 = source location of first ray
c---- tang1 = take-off angle in radians (positive angle implies
c---- in direction of increasing distance) of first ray
c----- sloc2 = source location of second ray
c---- tang2 = take-off angle in radians (positive angle implies
c-\cdots-- in direction of increasing distance) of second ray
c-\cdots-- datatime = actual travel time of configured ray
c---- xerr = Chuck Sword's statistic
c--..-
c---- nq = maximum number of rows in velocity grid
c-\cdots-- mq = maximum number of columns in velocity grid
c---- n = actual number of rows in velocity grid
c---- m = actual number of columns in velocity grid
c-\cdots-- nint = spatial increment between grid rows
c----- mint = spatial increment between grid columns
c-----
c\cdots--- vel(nq,mq)= values of velocity at grid points
c-----
c---- ray1travel( n*m,6) = array that stores cell,theta0,x,z,t for ray1's path
c---- num = number of points in ray1travel
c----- ray2travel(n*m,6) = array that stores cell,theta0,x,z,t for ray2's path
c---- num2 = number of points in ray2travel
c--.--
c---- raylocs = array containing locations of first ray for graphing
c----- raylocs2= array containing locations of second ray for graphing
c-----
c----- i,j,k= temporary integer variables
c---- t1,t2,t3,t4,t5= temporary double-precision variables
c--.--
c----- storage parameters:
c----- ray1x,ray2x = x values
c----- ray1z,ray2z = z values
c----- ray1time,ray2time = time values
c----- ray1theta0, ray2theta0 = theta0 values
c---- ray1tri(3,2),ray2tri(3,2) = tri values
c----- ray1triloc(3,2),ray2triloc(3,2) = triloc values
c----- ray1triflag,ray2triflag = triflag values
c-\cdots.--
c---- ray1exit = .true. if ray1 has exited the model area
c--.--
c---- oldtime = total time at last level
c----- oldxerr = xerr at last level
c--\cdots-- oldray1x }
c----- oldray2x
c---- oldz y--.-- parameters of last level; used in computations
c---- oldtime1 } for sade2
c----- oldtime2 }
c-----
```



```
    real*4 sloc1,tang1
    real*4 sloc2,tang2
    real*4 datatime,xerr
    real*4 nint,mint
    real*4 vel
    real*8 ray1x,ray2x,ray1z,ray2z
    real*8 ray1time,ray2time
    real*8 ray1triloc,ray2triloc
    real*8 ray1theta0,ray2theta0
    real*4 ray1travel,ray2travel
    real*8 oldtime,oldxerr
```

```
    real*8 oldray1x,oldray2x,oldz,oldtime1,oldtime2
    real*8 finalx1,finalx2,finalz,dtdz1,dxdz1,dtdz2,dxdz2
    real*8 finalv1,finalv2
    real*8 t1, t2, t3, t4, t5
    logical ray1triflag,ray2triflag
    logical ray1exit
    logical raybad
    integer*4 i,j,k
    integer*4 ray,ndat,nmax,mmax,max, dne
    integer*4 n,m,nq,mq,ndatq
    integer*4 ray1tri,ray2tri
    integer*4 num,num2
    integer*4 numnull,numturn,iterat
    parameter(nq=90,mq=210)
    parameter(ndatq=4200)
    dimension vel(nq,mq)
    dimension raybad(ndatq)
    dimension raylocs(2,nq*mq)
    dimension raylocs2(2,nq*mq)
    dimension ray1tri(3,2),ray2tri(3,2)
    dimension ray1triloc(3,2), ray2triloc(3,2)
    dimension ray1travel(nq*mq,6),ray2travel(nq*mq,6)
    common/velocity/vel
    common /iterat/iter
    cormon /travel/ray1travel,num,ray2travel,num2
    common /sade/finalx1, finalx2,finalz,dtdz1,dxdz1,dtdz2,dxdz2,
    finalv1,finalv2
    common /lgxerr/raybad
    common /nums/numnull,numturn
    cormon /conparm/n,m,nint,mint,nmax,mmax,max,ndat,dne
c
c----- set constants and initial variables
C
    t1=0.0d0
    t2=0.0d0
    t3=0.0d0
    t4=0.0d0
    t5=0.0d0
    ray1exit=.false.
    num=0
    num2=0
    do 10 i=1,n*m
        ray1travel(i,1)=0.0e0
        ray2travel(i,1)=0.0e0
        continue
10
c-\cdots.- put parameters for first ray into storage
c
    ray1x=dble(sloc1)
    ray1theta0=dble(tang1)
    ray1z=0.0d0
    ray1t ime=0.0d0
    ray1triflag=.true.
    ray1tri(1,1)=dint(dble(sloc1/mint))+1
    ray1tri(1,2)=1
    ray1tri(2,1)=ray1tri(1,1)+1
    ray1tri(2,2)=1
    ray1tri (3,1)=ray1tri(2,1)
    ray1tri(3,2)=2
    ray1triloc(1,1)=dble(ray1tri(1,1)-1)*dble(mint)
    ray1triloc(1,2)=0.0d0
    ray1triloc(2,1)=ray1triloc(1,1)+dble(mint)
    ray1triloc(2,2)=0.0d0
    ray1triloc(3,1)=ray1triloc(2,1)
    ray1triloc(3,2)=dble(nint)
c
c---- put parameters for second ray into storage
c
    ray2x=dble(sloc2)
```

```
    ray2theta0=dble(tang2)
    ray2z=0.0d0
ray2time=0.0d0
ray2triflag=.true.
ray2tri(1,1)=dint(dble(sloc2/mint))+1
ray2tri(1,2)=1
ray2tri(2,1)=ray2tri(1,1)+1
ray2tri(2,2)=1
ray2tri(3,1)=ray2tri(2,1)
ray2tri(3,2)=2
ray2triloc(1, 1)=dble(ray2tri(1,1)-1)*dble(mint)
ray2triloc(1,2)=0.0d0
ray2triloc(2,1)=ray2triloc(1,1)+dble(mint)
ray2triloc(2,2)=0.0d0
ray2triloc(3,1)=ray2triloc(2,1)
ray2triloc(3,2)=dble(nint)
c
c---- load initial points into graphing array and "raytravel's"
c
num=1
ray1travel(1,1)=float(ray1tri(1,1)*2+(ray1tri(1,2)-1)*(m-1)*2)
ray1travel(1,2)=tang1
ray1travel(1,3)=sloc1
ray1travel (1,4)=0.0e0
ray1travel(1,5)=0.0e0
ray1travel(1,6)=0.0e0
num2=1
ray2travel(1,1)=float(ray2tri(1,1)*2+(ray2tri(1,2)-1)*(m-1)*2)
ray2travel(1,2)=tang2
ray2travel(1,3)=sloc2
ray2travel(1,4)=0.0e0
ray2travel(1,5)=0.0e0
ray2travel(1,6)=0.0e0
c
c----- trace ray1 through to the next level
c
1000 oldtime=ray1time+ray2time
oldxerr=ray2x-ray1x
oldray1x=ray1x
oldray2x=ray2x
oldz=ray1z
oldtime1=ray1time
oldtime2=ray2time
11111 call raytrace(ray1x, ray1theta0,ray1z,ray1time,ray1triflag,
    + ray1tri,ray1triloc)
c
c----- load results into 'ray1travel' memory array and graphing array
c
    num=num+1
    if(num.gt.n*m) goto }8888
    i=0
    if(.not.ray1triflag) i=2*(m-1)+1
    ray1travel(num,1)=float(ray1tri(1,1)*2+(ray1tri(1,2)-1)*
    +
                                    (m-1)*2-i)
    ray1travel (num,2)=sngl(ray1 theta0)
        ray1travel (num,3)=sngl (ray1x)
        ray1travel (num,4)=sngl(ray1z)
        ray1travel (num,5)=sngl(ray1t ime)
        ray1travel(num,6)=0.0d0
c
c---- check to see if ray1 has left the model
c---- if not check to see if the next level has been reached
c
    if((ray1tri(1,1).lt.1).or.(ray1tri(2,1).gt.m).or.
    + (ray1tri(3,2).lt.1)) then
            ray1exit=.true.
            xerr=0.0e0
            return
            endif
    if(ray1z.le.ray1travel(num-1,4)) then
            numturn=numturn+1
```

```
            xerr=0.0e0
            return
            endif
        if(dmod(ray1z,dble(nint)).ne.0.0d0) goto 11111
c
c----- trace ray2 through to the next level
c
22222 call raytrace(ray 2x, ray2theta0, ray 2z,ray2time,ray2triflag,
    + ray2tri,ray2triloc)
c
c----- load results into 'ray2travel' memory array and graphing array
c
        num2=num2+1
        if(num2.gt.n*m) goto }8888
        i=0
        if(.not.ray2triflag) i=2*(m-1)+1
        ray2travel(num2,1)=float(ray2tri(1,1)*2+(ray2tri(1,2)-1)*
    + (m-1)*2-i)
        ray2travel(num2,2)=sngl (ray2theta0)
        ray2travel(num2,3)=sngl(ray2x)
        ray2travel (num2,4)=sngl (ray2z)
        ray2travel(num2,5)=sngl(ray2time)
        ray2travel(num2,6)=0.0d0
c
c----- check to see if ray2 has left the model
c----- if not check to see if the next level has been reached
c
    if((ray2tri(1,1).lt.1).or.(ray2tri(2,1).gt.m).or.
            (ray2tri(3,2).(t.1)) then
            xerr=0.0e0
            return
            endif
    if(ray2z.le.ray2travel(num2-1,4)) then
        numturn=numturn+1
        xerr=0.0e0
        return
        endif
    if(dmod(ray2z,dble(nint)).ne.0.0d0) goto 22222
    if(ray1exit) goto 99999
c
c----- check to see if the data time has been surpassed
    if((ray1time+ray2time).ge.dble(datatime)) goto 99999
c---- check to see if the last level has been reached
c----- otherwise loop back to trace to next level
c
    if(idnint(ray1z/dble(nint)).eq.(n-1)) then
        xerr=0.0e0
        return
        endif
    goto 1000
C
c----- if n*m intersection points exists, write to screen and output zero
c
88888 write(2,'(2x,a40)') '******* number of int points exceeded '
    xerr=0.0
    return
c
c----- calculate 'xerr' (linearly interpolating between levels)
c
99999 t1=ray2x-ray1x
    t2=ray1time+ray2time
    t3=((dble(datat ime)-oldtime)/(t2-oldt ime))
    t4=t3*(t1-oldxerr)
    xerr=sngl(oldxerr+t4)
    if(abs(xerr).gt.350.0) then
    xerr=0.0
    raybad(ray)=.true.
        numnul l=numnul l+1
        return
```

endif
c
compute $z$, final $x$ 's and the four derivatives
C
final $x 1=0$ ldray $1 x+($ ray $1 x$-oldray $1 x$ )*t 3
final $x 2=$ oldray $2 x+($ ray $2 x$-oldray $2 x$ )*t3
final $z=01 d z+t 3^{*} d b l e(n i n t)$
$d t d z 1=($ ray1time-oldtime1)/nint
$d x d z 1=($ final 1 1-oldray $1 x) /\left(t 3^{*}\right.$ nint $)$
dtdz2=(ray2time-oldtime2)/nint
$d x d z 2=($ final $x 2$-oldray $2 x) /\left(t 3^{*}\right.$ nint $)$
finalv1=dmax1(dble(vel(ray1tri(1,2), ray1tri(1,1))), dble(vel(ray1tri(2,2), ray1tri(2,1))),
dble(vel(ray1tri(3,2), ray1tri(3,1))))
$+\quad$ dble(vel(ray1tri(3,2), ray1tri(3,1))))
finalv2 $=$ dmax1 (dble(vel(ray2tri(1,2),
dble(vel(ray2tri(2,2),
ray2tri $(2,1)))$,
$+$
dble(vel(ray2tri(3,2), ray2tri(3,1))))

## return

end


[^0]:    ${ }^{1}$ Since the synthetic data is noise free, the noise in the solution stems from the underdetermined nature of the inverse problem.

[^1]:    ${ }^{1}$ There will be more discussion on the required size of this threshold in a later section.
    ${ }^{2}$ The velocity at the correct centre must be zero, so it is easy to check the two choices.

[^2]:    ${ }^{3}$ One needs to be mindful of the direction of energy transfer around the circle, therefore being able to assign a $360^{\circ}$ range of angles.

[^3]:    ${ }^{4}$ The observations made about asymmetry here are reinforced by Bregman et al., 1989. Later in this thesis (Figures 2.2-4, 2.2-5 and 2.5-1) some synthetic examples display significant asymmetry even though the model and raypaths appear symmetric. This is not a result of the triangular cell shape. The model traveltime data were derived with exactly the same raytracing code and triangular cells. That is, the models themselves effectively have an asymmetric velocity distribution even though the structure of the anomalous velocity nodes is symmetric. The raypaths used for these examples are symmetrical with respect to the extent of the model, however, this means that they sample the edges of the (asymmetric) anomaly differently. The asymmetry of the solution is a result of this sampling and the subsequent response of the inversion algorithm to the underdetermined problem.

[^4]:    ${ }^{1}$ Although Jacobs et al., 1992, and Lailly et al., 1992, suggest inverse raytracing of the picked migrated data to predict the pre-migration traveltimes.

[^5]:    ${ }^{1}$ Including diffracted events, fault-plane reflections and multiple arrivals.
    ${ }^{2}$ The ray parameter is the inverse of apparent horizontal velocity.

[^6]:    ${ }^{3}$ The unit step function is zero for $\omega<0$, unity for $\omega>0$ and a $1 / 2$ for $\omega=0$.

[^7]:    ${ }^{5}$ Remember that the negative frequencies were zeroed.

[^8]:    ${ }^{6}$ Scheuer and Oldenburg (1988) preferred velocity filtering for their applications. Velocity filtering is avoided here because it totally removes steeply dipping energy that may provide some useful information. High cut filtering provides the opportunity for such energy to be preserved. Velocity filtering is also avoided as it requires the added computational expense of a 2D FFT whereas high cut filtering requires only 1D FFT's.

[^9]:    ${ }^{2}$ Called semi-analytic because only part of the ray is retraced. The rest is an analytical approximation.

[^10]:    ${ }^{3}$ On an IBM 3090 mainframe computer, with vector facility.

[^11]:    ${ }^{1}$ For more on MEMNL\$8, see section 4.

[^12]:    ${ }^{2}$ Many of these rays exited the boundaries of the model. These were ignored and another ray traced until 800 valid rays were left.

[^13]:    ${ }^{3}$ This anomaly has a velocity that is $25 \%$ higher than the velocity of its host layer.

[^14]:    ${ }^{4}$ ' $n$ ' is the number of velocity nodes in the grid.

[^15]:    ${ }^{2}$ Carrion (1991) has proposed a tomography scheme based on the assignment of a Lagrange multiplier for each ray.
    ${ }^{1}$ This is the general constraint statistic - the traveltime residual. In this study the $X_{e r m(i)}$ statistic is used (see sections 2.2 and 2.4).

[^16]:    ${ }^{1}$ Also known as the "minimum norm" solution.

[^17]:    ${ }^{1}$ It is possible to construct a single function, $H$, using a Lagrange multiplier, $\lambda$ (ie. $H-C-\lambda S$ ). Skilling and Bryan (1984) claim that simultaneous optimisation is easier because of complications in finding the required value of $\lambda$.

[^18]:    ${ }^{5} D_{\beta}=\operatorname{diag}\left[\beta_{1}, \beta_{2}, \ldots\right]$ where $\beta_{1}, \beta_{2}, \ldots$ are the eigenvalues of $B$. with respect tox

[^19]:    ${ }^{2}$ Other proportions were tested in this study but this specification generally gave the best results.

[^20]:    ${ }^{3}$ It is possible for the maximum entropy point to be multi-valued (see Skilling and Bryan, 1984). This possibility does not create any problems though.

[^21]:    ${ }^{2} \mathrm{I}$ is the $(r \times r)$-identity matrix.

[^22]:    ${ }^{1}$ The second term on the right of this equation disappears when $\nabla C$ is evaluated at the current model, $\mathrm{v}_{0}$, at each step of the iterative inversion.

[^23]:    ${ }^{1}$ Skilling and Bryan, 1984, advocated TEST $<0.1$ in their applications.

[^24]:    ${ }^{1}$ These will be harder to image in any case due to the surface restrictions of the raypaths.

[^25]:    ${ }^{1}$ Cable feathering would cause some complications.

