# Discontinuous Galerkin Time Domain Methods for Acoustics and Comparison with Finite Difference Time Domain Methods 

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#### Abstract

This thesis describes an implementation of the discontinuous Galerkin finite element time domain (DGTD) method on unstructured meshes to solve acoustic wave equations in discontinuous media. In oil industry people use finite difference time domain (FDTD) methods to compute solutions of time domain wave equations and simulate seismic surveys, the first step to explore oil and gas in the earth's subsurface, conducted either in land or sea. The results in this thesis indicate that the first order time shift effect resulting from misalignment between numerical meshes and material interfaces in the DGTD method occurs the same way as interface errors in the finite difference simulation of wave propagation. This thesis describes two approaches: interface-fitting mesh and local mesh refinement, without modifying the DGTD scheme, to decrease this troublesome effect with verifications of 2D examples. The comparison in this thesis between the DGTD method on the piecewise linear interface-fitting mesh and the staggered FDTD method both applied to a square-circle model and a 2D dome model confirms the fact that the DGTD method can achieve a second order convergence rate while the error in the staggered FDTD method is dominated by the first order interface error. I end with the conclusion that the DGTD method requires less computation cost than the staggered FDTD method for the two solutions to have roughly the same accuracy for the more realistic 2 D dome model.


## INTRODUCTION CHAPTER

The numerical solutions of wave propagation have numerous applications. One of particular interest and importance is to simulate seismic surveys in oil and gas exploration. Various numerical methods have been developed to carry out this simulation. In this thesis I describe an implementation of the discontinuous Galerkin finite element time domain method (DGTD) applied to acoustic wave equations in heterogeneous media and use computer programs I developed to evaluate its accuracy and efficiency. Furthermore, I will make comparison between DGTD and finite difference time domain method, a fully-developed method used daily as the wave propagator in reflection seismology for decades. This comparison may shed light on people's understanding of benefits and drawbacks of the two methods.

## Motivation

The first step in oil and gas exploration either in land or sea is to conduct seismic surveys, which typically consists in sending into the ground sound waves generated by sources at the surface such as air guns in marine surveys or dynamite in land acquisition and through sensors called geophones recording echoes of the sending waves, caused by the heterogeneity of the earth's subsurface. Seismologists and geoscientists then can analyze the recorded time series of data called seismic traces or seismograms and interpret the earth's interior properties by imaging technologies based on the basic mathematical point of view that "waves transfer space-time resolved information from one place to another with (relatively) little loss" (Symes, 2003).

Numerical seismic modeling, duplicating the seismic survey procedure and generating synthetic seismograms provided the earth's subsurface structure, has appli-
cations at least in two aspects including the inversion process and the experimental design. Reflection seismic inversion problem is usually set to find the geological model given observed seismic data. For example, in linearized inversion problems the oscillatory part of the geological model is updated iteratively by minimizing an objective function, for example, a distance function related to observed data (seismograms) and synthetic data. Since each iteration a wave propagation problem with highly intensive computation must be solved and thousands of iterations are required, an efficient and accurate wave propagation solver is highly demanded in this context.

In addition, numerical modeling has been used for designs of acquisition geometries. Source and receiver geometries decide the subsurface image quality in the later process after seismic surveys. Regone (2007) used 3D finite difference modeling, instead of unfeasible 3D field surveys due to the cost, to convince people the great improvement in image quality with wide-azimuth surveys over traditional narrowazimuth towed-streamer surveys, and came up with two newly acquisition geometries: receivers distributed in a sparse grid on the ocean floor with sources distributed in a dense grid on the surface (OBS) and multiple vessels wide-azimuth towed-streamer surveys (WATS), which both have been used in practice.

Various numerical methods are applicable to wave propagation problems. Finite difference (FD) and discontinuous Galerkin (DG) methods are of this thesis's concerns. FD methods become an industry standard for solving wave propagation problems due to their desirable trade-off between the computation efficiency and accuracy, as well as the relatively easy implementation of FD methods. Discontinuous Galerkin (DG) methods draw a lot of attentions in computational electromagnetic and fluid dynamics communities recently since DG methods specialize in solving hyperbolic partial differential equation (PDE) and dealing with complex geometries. Because (a) seismic wave equations are one class of the most basic linear hyperbolic PDEs, (b) the heterogeneity of geological model and the irregular surface landform
in land acquisition are in need of discrete grids having the flexibility to represent complex structures precisely, DG methods are a competitive candidate for seismic wave simulation.

Though a lot of efforts have been paid on both methods respectively, there are few works regarding careful comparison between them. This work as intended is to study this aspect within the context of reflection seismology, and come up with some useful conclusions which may provide a guideline of choosing the proper method for the proper problem.

In this work I choose the acoustic wave equations (pressure-velocity formulation) to which the two methods are applied. The acoustic wave equations form a linear PDE system in which the acoustic pressure and particle velocity interact with one another, according to Hooke's law and conservation of momentum, to propagate acoustic waves through materials. Although elastic wave simulation is more close to the way waves propagate in the subsurface, acoustic modeling is good enough for this study for several reasons.

First, material parameters are easier to prepare for acoustic modeling. Material density and bulk modulus, or sound speed, are all needed for acoustic modeling, while in elastic modeling the stress-strain relation itself can asks for up to 21 independent material parameters, some of which are still mysterious. Even in the isotropic case, how to interpret shear velocity near the ocean bottom is still a mystery to geoscientists. Second, one can obtain analytic solutions of acoustic wave equations in some cases. Therefore I can carry out conventional convergence tests rather than the ones through error estimations. Third, for free surface boundary, absorbing boundary, ocean floor nearby and so on, the two methods in elastic modeling have their own delicate numerical treatments, which may render the comparison less objective. Acoustic wave equations as a simplified model don't have these tedious and tricky
implementation issues. The comparison for acoustic modeling can reveal underlying intrinsic natures of the two methods themselves as much as possible. Last but not least, acoustic modeling has been used as grounds for many of processing technologies. In future work, I may use acoustic modeling based on DGTD methods as a built-in part of imaging process and evaluate it at the next level of applications in reflection seismology.

## Review

Characterizations of convection phenomenons are illustrated through natural and artificial events from the hurricane formation to the oil and gas production. A class of hyperbolic PDEs under certain mathematical assumptions give descriptions of the evolution of these phenomenons. Toro (1997) from both theoretical and numerical aspects give a review of hyperbolic PDEs originating from fluid dynamics. Mathematical derivations of both acoustic and elastic wave equations of this thesis's interest can be seen in Leveque (2002), which also includes theories and numerical methods (finite volume methods) for general hyperbolic problems. Bedford and Drumheller (1994) offers a basic introduction to linear elasticity for wide audience. The first part of Cohen (2002) also provides a good reference for understanding basic definitions and properties of wave equations.

DG methods have been applied to a wide range of hyperbolic problems, including gas and fluid dynamics (Cockburn and Shu, 1989; Bassi and Rebay, 1997; Giraldo et al., 2002), Maxwell's equations (Warburton, 1999; Hesthaven and Warburton, 2002; Cockburn et al., 2004; Cohen et al., 2006), acoustic and elastic wave equations (Atkins and Shu, 1998; Käser and Dumbser, 2006) and so on for more than three decades, since the first DG scheme was introduced by Reed and Hill (1973) for solving a neutron transport equation. DG methods can be viewed as an extension of finite vol-
ume (FV) methods in the sense that DG methods allow discontinuous approximation of solutions. Techniques developed for FV methods fit into DG methods naturally (Cockburn et al., 1999). For example, numerical fluxes, through which underlying physical laws could be imposed into the numerical scheme so as to ensure the numerical stability, have been studies for a while by authors (Toro, 1997; Leveque, 2002), whose results can be directly used in the DG framework. Hesthaven and Warburton (2008) scrutinize DG methods together with other popular numerical methods for PDEs and outline the general properties of DG methods and conclude that DG methods possess every useful feature discussed. Cockburn et al. (1999) and Hesthaven and Warburton (2008) list generic properties of DG methods as follows,

- DG methods are good at dealing with complex geometries and usually form easy implementations for boundary conditions.
- For time dependent PDEs, DG methods can formulate explicit semi-discrete form due to the locally defined mass matrix. This avoids inverting a large-scale mass matrix such as in conventional finite element methods. high order schemes
- High order accuracy under DG framework is achievable for the problems whose exact solutions are smooth. This is a good news for this work because solutions of wave equations usually have certain smoothness property.
- Implementation of hp-adaptivity in DG methods is less troublesome than conforming finite element methods, because no continuity restrictions are imposed on numerical solutions.

The convergence analysis of DG methods, started by Lesaint and Raviart (1974) assuming analytic solutions are smooth, proved $h^{p+\epsilon}$ convergence rate, where $h$ is grid size, $p$ is the degree of local basis polynomials and the value $\epsilon$ is determined by ways of the triangulation, for example, 0 for general grids and 1 for Cartesian grids. Johnson
and Pitkaranta (1986) later ameliorated $\epsilon$ to be $1 / 2$ for general grids. Ainsworth (2004) studied the dispersive behavior of DG methods and showed an exponential decay of the relative phase error provided $2 p+1 \approx c k h$ for some fixed constant $c>1$, where $k$ is spatial wave-number. This result is critical because it states a rule for choosing the order of basis functions with respect to the grid size so as to achieve the expected error in wave simulations with DG methods. A similar criterion for finite-difference acoustic modeling can be found in Alford et al. (1974).

For time dependent problems, a proper time integration method, i.e., an ODE solver, is required after the spatial discretization with DG methods. To avoid solving large-scale linear systems each time step, an explicit ODE solver is preferred. Chavent and Salzano (1982) formed a DG method for spatial discretization in 1D scalar conservation law and used a forward Euler method for time integration. This method by a numerical analysis, however, is shown to be stable only if $\Delta t \sim \mathcal{O}\left(\Delta x^{3 / 2}\right)$, which implies unacceptably small time step. The high-order accurate Runge-Kutta (RK) DG methods, generalized by Cockburn and Shu (1989), are widely used because they match the temporal and spatial discretization accuracy, and just require $\Delta t \sim \mathcal{O}(\Delta x)$. The ADER-DG methods in Käser and Dumbser (2006) are also an interesting way to achieve high order accuracy in both space and time by using arbitrary high order derivatives to construct time discretization. This work uses the low-storage five-stage fourth-order explicit RK method for time discretization in the DG implementation as discussed in Hesthaven and Warburton (2008).

Finite difference (FD) methods becomes a robust tool and an industry standard in seismic wave simulation, because FD methods perform a good balance between accuracy and efficiency and are easily implemented. Owing to the efforts of many authors (Richtmyer and Morton, 1967; Mitchell and Griffiths, 1994; Cohen, 2002), basic issues related to FD methods, such as consistency, stability and convergence, are well studied and criterion for designing FD methods for different problems are
provided. One can expect reliable and accurate solutions from a suitable FD method.

In FD methods, space and time are discretized by grids, where field variables as well as material parameters are defined. Differential operators are approximated by finite-difference formulas. In seismic wave simulation, the discrete grids are usually distributed equally on a Cartesian coordinate system, though other systems are possible. The benefits of using such grids are explicit grid-neighbor relation and efficient memory access pattern.

There are two approaches of FD methods, conventional-grid and staggered-grid approaches differentiated with each other in the way whether unknowns of field variables and/or material parameters are defined on the same grid or not. In the 1960s and 1970s, conventional-grid approach was popular for the displacement formulation of elastic wave equations, of which functions are approximated at the same discrete grid (Alford et al., 1974; Alterman and Karal, 1968; Boore, 1970, 1972; Dablain, 1986; Kelly et al., 1976). Unfortunately, grid dispersion and numerical instabilities when material parameters has high contrast discontinuities influence numerical solutions badly in this approach. Staggered-grid approach, in which several grids are employed, overcomes these difficulties and so are widely used recently. Yee (1966) first applied a staggered-grid second order FD scheme to Maxwell's equations in isotropic media. Madariaga (1976) later used a second order staggered-grid FD scheme to model the earthquake rupture. Virieux $(1984,1986)$ formally established the staggered-grid approach to solve velocity stress formulation of elastic wave equations. The fourthorder staggered-grid FD method introduced by Bayliss et al. (1986) was then proved to be more efficient and require less memory than the second-order method, since the grid interval in the fourth-order method can be twice as much as the one in the second-order method according to the rule of thumb proposed by Alford et al. (1974).

High-order FD time domain methods are constructed to control the grid dispersion
effectively for models with smooth parameters. However, FD methods for seismic models with high contrast discontinuities of material parameters lead to a first-order interface error, which is irrelevant to FD methods, but stems from the insufficient representation of model parameters on discrete grids. Brown (1984) first analyzed this first order error component of FD methods for an interface problem. Symes and Vdovina (2009) theoretically and numerically quantify the first order interface misalignment error for the second order in time and space staggered-grid FD method applied to the pressure-velocity formulation of the acoustic wave equation, and provide an explicit expression of a non-zero time shift of numerical solutions due to the first order interface error. In their discussion, the interface error in staggered-grid FD method is unavoidable for heterogeneous media, because several grids are employed in staggered-grid FD method and misalignment with the material interface must occur for at least one computational grid.

## Claim

This thesis through numerical examples demonstrates DG methods somehow can remedy this interface error by using the piecewise linear interface-fitting mesh and local mesh refinement. Given a mesh of elements with straight-line edges (triangles for example), each element can be considered as a homogeneous media. In this way, one actually glues piecewise constant medium in each element together to approximate the real model. Interface-fitting meshes align the vertices of elements with material interfaces. Provided the seismic model can be detached into several homogeneous media and interface-fitting mesh for this model is generated, (a) when interfaces are composed of line segments, the model approximation on this mesh is exact, so the numerical error comes from DG methods only; (b) when interfaces are curved and this mesh is not too coarse, the model approximation brings in a second order error
since line segments of elements can not fully represent the curved interfaces, therefore the second order convergence rate is the best one can expect in this case. In short, by using interface-fitting meshes, the second order convergence rate is achievable with DG methods on interface-fitting meshes. However this kind of meshes sometimes are hard to generate especially when the material structures are complicated.

Local mesh refinement is another approach proposed in this thesis to decrease the interface error. By locally refining the mesh near the interface, the interface error is reduced as the distance between the interface and the nearest grid points is shortened. The drawback of this technique is that one can get very small time step due to the small elements by the refinement.

I compares the computation cost, measured by the wall clock time or the number of float point operations, for the two solutions by FD and DG methods to have roughly the same accuracy. For the heterogeneous model with piece-wise constant media, DG methods on a interface-fitting mesh can achieve second order convergence rate while the numerical error of staggered-grid FD methods is dominated by the first order interface error. In this case, DG methods are more efficient for a prescribed accuracy ( $5 \%$ for example). This conjecture is demonstrated by numerical examples.

I also implement the low-storage curvilinear DG method. In this method curvilinear elements instead of straight sided elements are used to approximate the geometry of the model, such as the material interfaces or boundaries. In this way the geometry representation by curvilinear elements complements the accuracy of the DG solver. We can expect the optimal convergence rate for the numerical simulation. Compared to the 2 nd order convergence rate on straight sided elements, this method is more tempting. In order to form the curvilinear elements, one need the geometry information as precisely as possible. But the exact expression of the geometry for a realistic model may be unknown. It's impossible to built curvilinear elements and hence the
curvilinear DG method is not applicable.

## Agenda

This thesis is organized as follows. First the model problem, i.e., pressure-velocity formulation of the acoustic wave equations, is introduced. For the model problem, analytic solutions of three special cases are discussed. The first two cases are used in the convergence tests later. The last one, called Riemann problem, showing a way to construct numerical flux for the linear system PDE, is recalled when the DG method is formulated. Then I construct the DG method following Hesthaven and Warburton (2008). Later, two types of boundary conditions widely used in seismic simulation are discussed. At the end of the method chapter, I present the 2-4 staggered-grid FDTD method for acoustics used for the comparison.

Several numerical experiments are carried out for the convergence tests of DGTD methods and comparison of DGTD and FDTD. I first present the point source wave and plane wave experiments for the purpose of the convergence tests. Then the interface error in DGTD methods is illustrated by using the misaligned mesh, while the interface-fitting mesh and local refined mesh examples show the approaches at mesh level in DGTD methods to reduce this unpleasant error component. At last I make comparison of the two methods with respect to the computation cost and the numerical accuracy. The square-circle model and 2D dome model are used for such comparison.

After numerical experiments, I'll discuss the formulation of the low-storage curvilinear DG method. Then this method is applied to the square-circle model and 2D dome model. The curvilinear DG method as shown in the numerical examples achieves the optimal convergence rate.

## METHODS CHAPTER

## Introduction

This chapter first introduces the pressure-velocity formulation of the acoustic wave equations, for which three special cases are discussed. The analytic solutions of the first two cases are provided for the convergence tests. The last case forms a Riemann problem. The process for the construction of its solution shows a way to formulate the numerical flux for the acoustic wave equations in DG methods. Next, the DG discretization for AWE in space is built step by step following Hesthaven and Warburton (2008). I omit basic theoretical discussions, but focus on presenting a self-sustained construction procedure, with which one can develop and implement a DG method by himself/herself. Finally, Richardson extrapolation is discussed for convergence rate estimation in the case where the analytic solutions are inaccessible.

## Model problem

The pressure-velocity formulation of the acoustic wave equations (AWE) is expressed as a linear PDE system, in which the acoustic pressure and particle velocity interact with one another to propagate waves through materials. The purely hyperbolic property makes such system a good prototype (starting point) to understand the behaviors of the hyperbolic PDEs and develop effective and efficient numerical methods.

This first-order linear system in term of pressure $p$ and velocity reads,

$$
\begin{array}{r}
\rho(\mathbf{x}) \frac{\partial \mathbf{v}}{\partial t}+\nabla p=0 \\
\frac{1}{\kappa(\mathbf{x})} \frac{\partial p}{\partial t}+\nabla \cdot \mathbf{v}=f\left(\mathbf{x}, t ; \mathbf{x}_{s}\right) \tag{1}
\end{array}
$$

where $f$ represents source density function with respect to source location $\mathbf{x}_{s} \in \mathbb{R}^{n}$ ( $n=1,2$, or 3 ) as well as space $\mathbf{x} \in \mathbb{R}^{n}$ and time $t ; \rho$ and $\kappa$ denote the mass density and bulk modulus, respectively, which are dependent of space $\mathbf{x}$ only; speed of the wave is $c(\mathbf{x})=\sqrt{\frac{\kappa(\mathbf{x})}{\rho(\mathbf{x})}}$ and acoustic impedance is $Z(\mathbf{x})=\sqrt{\kappa(\mathbf{x}) \rho(\mathbf{x})}$

## Point source wave

Ideally, source density function $f$ is viewed as a isotropic point radiator with a known time-dependent function as,

$$
\begin{equation*}
f\left(\mathbf{x}, t ; \mathbf{x}_{s}\right)=w(t) \delta\left(\mathbf{x}-\mathbf{x}_{s}\right), \tag{2}
\end{equation*}
$$

which is a quite coarse approximation to the real source function. However, it is fairly good for this thesis. Usually, the source pulse $w(t)$ is of compact support in time, that is $w(t)=0,|t|>t_{0}$. This mimics the acoustic energy generated by airguns and dynamite during a limited time. To simply see how waves expand starting from a point, I hereby assume the media is homogeneous, that is $\rho(\mathbf{x})=\rho_{0}$ and $\kappa(\mathbf{x})=\kappa_{0}$, $\forall \mathbf{x}$. The model problem Eqs.(1) then are equivalent to the second order wave equation for the acoustic potential $u$ with proper initial and boundary conditions,

$$
\begin{equation*}
\frac{1}{\kappa_{0}} \frac{\partial^{2} u}{\partial t^{2}}-\nabla \cdot \frac{1}{\rho_{0}} \nabla u=w(t) \delta\left(\mathbf{x}-\mathbf{x}_{s}\right) \tag{3}
\end{equation*}
$$

where $u(\S, t)=\int_{-\infty}^{t} p(\mathbf{x}, s) d s$, thereby $p=\frac{\partial u}{\partial t}$ and $\mathbf{v}=-\frac{1}{\rho_{0}} \nabla u$. The solution of Eq.(3) with vanishing initial conditions describes outgoing spherical waves at speed $c_{0}$ as

$$
\begin{equation*}
u(\mathbf{x}, t)=\rho_{0} \frac{w\left(t-r / c_{0}\right)}{4 \pi r}, \quad r=\left|\mathbf{x}-\mathbf{x}_{s}\right| . \tag{4}
\end{equation*}
$$

## Two-layer media

The material parameters $\rho$ and $\kappa$ are usually non-constants, but vary with location. Due to the material discontinuity, or called material interface, the incident wave is decomposed into a transmitted wave and a reflected wave. Two-layer medium is a good simplified model to replicate this wave decomposition procedure. In two-layer medium, the material parameters have a jump at $x=0$, but keep invariant with respect to other spatial variables,

$$
(\kappa(x), \rho(x))= \begin{cases}\left(\kappa_{l}, \rho_{l}\right), & x \leq 0  \tag{5}\\ \left(\kappa_{r}, \rho_{r}\right), & x>0\end{cases}
$$

Eqs.(1) without source can be further simplified into 1D case, because of field variables only dependent of $x$,

$$
\begin{align*}
\rho(x) \frac{\partial v(x, t)}{\partial t} & =-\frac{\partial p(x, t)}{\partial x} \\
\frac{1}{\kappa(x)} \frac{\partial p(x, t)}{\partial t} & =-\frac{\partial v(x, t)}{\partial x} \tag{6}
\end{align*}
$$

Assuming the incident wave impacts on the interface from the left $(x<0)$, the solution of Eqs.(6) with continuity restriction at the interface $x=0$ is then given as,

$$
\begin{align*}
& p(x, t)=\left\{\begin{array}{cc}
g\left(t-x / c_{l}\right)-\frac{\rho_{l} c_{l}-\rho_{r} c_{r}}{\rho_{l} c_{l}+\rho_{r} c_{r}} g\left(t+x / c_{l}\right), & x \leq 0 \\
\frac{2 \rho_{r} c_{r}}{\rho_{l} c_{l}+\rho_{r} c_{r}} g\left(t-x / c_{r}\right), & x>0
\end{array}\right. \\
& v(x, t)=\left\{\begin{array}{cc}
\frac{1}{\rho_{l} c_{l}}\left(g\left(t-x / c_{l}\right)+\frac{\rho_{l} c_{l}-\rho_{r} c_{r}}{\rho_{l} c_{l}+\rho_{r} c_{r}} g\left(t+x / c_{l}\right)\right), & x \leq 0 \\
\frac{2}{\rho_{l} c_{l}+\rho_{r} c_{r}} g\left(t-x / c_{r}\right), & x>0
\end{array}\right. \tag{7}
\end{align*}
$$

where $g$ is a continuous function compatible with the initial condition.

Two-layer medium model as well as its solution set an interesting and practicable
example to study the error components in numerical methods, see Symes and Vdovina (2009). I will use the same example to illustrate the error components in DG methods later and propose approaches to suppress the interface error as much as possible.

## Riemann problem

Previously I have studied the cases with continuous solutions due to the continuous initial conditions. Now I consider the problem with piece-wise constant initial data, called Riemann problem. I start with 1D acoustic wave equations Eqs.(6) in a constant medium and construct its solution. Then I extend to the heterogeneous media and multidimensional cases. The discussion below is partly adopted from Leveque (2002).

Eqs.(6) in a constant medium with piecewise constant initial data can be rewritten into matrix-vector form as,

$$
\begin{array}{r}
\frac{\partial \mathbf{q}}{\partial t}+A(x) \frac{\partial \mathbf{q}}{\partial x}=0,  \tag{8}\\
\mathbf{q}(x, 0)= \begin{cases}\mathbf{q}_{L}, & x<0 \\
\mathbf{q}_{R}, & x>0\end{cases}
\end{array}
$$

with

$$
\mathbf{q}=\left[\begin{array}{l}
v(x, t) \\
p(x, t)
\end{array}\right], \quad \mathbf{q}_{L}=\left[\begin{array}{l}
v_{L} \\
p_{L}
\end{array}\right], \quad \mathbf{q}_{L}=\left[\begin{array}{l}
v_{R} \\
p_{R}
\end{array}\right], \quad A(x)=\left[\begin{array}{cc}
0 & \kappa_{0} \\
1 / \rho_{0} & 0
\end{array}\right],
$$

where $A$ is a constant matrix and diagonalizable, that is $A=R \Lambda R^{-1}$ with

$$
R=\left[r_{1}, r_{2}\right]=\left[\begin{array}{cc}
-\sqrt{\rho_{0} \kappa_{0}} & \sqrt{\rho_{0} \kappa_{0}}  \tag{9}\\
1 & 1
\end{array}\right], \Lambda=\left[\begin{array}{cc}
-c_{0} & 0 \\
0 & c_{0}
\end{array}\right] .
$$

By introducing characteristic variables $W=R^{-1} Q=\left[w_{1}, w_{2}\right]^{T}$ and multiplying Eq.(8)
by $R^{-1}$, it is then deduced into two decoupled advection equations,

$$
\begin{align*}
& \frac{\partial w_{1}}{\partial t}-c_{0} \frac{\partial w_{1}}{\partial x}=0 \\
& \frac{\partial w_{2}}{\partial t}+c_{0} \frac{\partial w_{2}}{\partial x}=0 \tag{10}
\end{align*}
$$

with piecewise constant initial conditions,

$$
W(x, 0)= \begin{cases}R^{-1} \mathbf{q}_{L}:=\left[w_{1, l}, w_{2, l}\right]^{T}, & x<0 \\ R^{-1} \mathbf{q}_{R}:=\left[w_{1, r}, w_{2, r}\right]^{T}, & x>0\end{cases}
$$

whose solutions are

$$
\begin{aligned}
& w_{1}(x, t)=w_{1}\left(x+c_{0} t, 0\right)= \begin{cases}w_{1, l}, & x+c_{0} t<0 \\
w_{1, r}, & x+c_{0} t>0\end{cases} \\
& w_{2}(x, t)=w_{2}\left(x-c_{0} t, 0\right)= \begin{cases}w_{2, l}, & x-c_{0} t<0 \\
w_{2, r}, & x-c_{0} t>0\end{cases}
\end{aligned}
$$

Finally, I can recovery the solution of Eq.(8) as a linear combination of the right eigenvectors $r_{1}, r_{2}$ of $A$.,

$$
\begin{align*}
\mathbf{q}(x, t) & =R W(x, t)=r_{1} w_{1}(x, t)+r_{2} w_{2}(x, t)  \tag{11}\\
& =\left\{\begin{array}{cc}
r_{1} w_{1, l}+r_{2} w_{2, l}, & x<-c_{0} t \\
r_{1} w_{1, r}+r_{2} w_{2, l}, & -c_{0} t<x<c_{0} t \\
r_{1} w_{1, r}+r_{2} w_{2, r}, & x>c_{0} t
\end{array}\right.
\end{align*}
$$

The two characteristic lines $x= \pm c_{0} t$ divide the upper-half $x-t$ plane into three regions. In each region, the solution $\mathbf{q}(x, t)$ is constant build up with different superposition of $r_{1}, r_{2}$.

One thing to notice is that the above derivation is valid not only for constant


Figure 1: Solution of Riemann problem in two-layer media in $x-t$ plane has three states separated by the two characteristic lines $x=-c_{l} t$ and $x=c_{r} t$.
medium, but also for cases where $R^{-1} \frac{\partial \mathbf{q}}{\partial x}=\frac{\partial\left(R^{-1} \mathbf{q}\right)}{\partial x}$, i.e., $R$ is independent of $x$. This is true when the left and right acoustic impedance's equal, i.e., $Z_{l}=Z_{r}$ for two-layer medium ( $c_{l}$ and $c_{r}$ are different). The solution of this case has the same structure as Eqs.(11), but the left and right wave speeds are different.

If $R(x)$ depends on $x$ then $R^{-1}(x) \frac{\partial \mathbf{q}(x, t)}{\partial x}=\frac{\partial\left(R^{-1}(x) \mathbf{q}(x, t)\right)}{\partial x}-\frac{d R^{-1}(x)}{d x} \mathbf{q}(x, t)$. The extra term $\frac{d R^{-1}(x)}{d x} \mathbf{q}(x, t)$ as a source term ruins the independence of the leftgoing and right-going waves and couples them together. Thus the solutions are complicated compared to the constant impedance case.

Leveque (2002) suggest to solve a general Riemann problem at a interface between two different materials. As indicated in Fig.(1), the two characteristic lines $x=-c_{l} t$ and $x=c_{r} t$ cut the upper half $x-t$ plane into three regions. From Eqs.(11), it is seen that the solution of Riemann problem is a linear combination of two waves represented by the two right eigenvectors of the material matrix $A$. Since the waves propagate the initial data to the left region $\left(x<-c_{l} t\right)$ and the right region $\left(x>c_{r} t\right)$, $\mathbf{q}^{l}(x, t)=\mathbf{q}_{L}$ and $\mathbf{q}^{r}(x, t)=\mathbf{q}_{R}$. The intermediate state $\mathbf{q}^{m}(x, t)$ fill in the middle region $\left(-c_{l} t<x<c_{r} t\right)$ left by the left-going and right-going waves. We also notice
that the solution jump across the two characteristic lines is parallel with an right eigenvector of coefficient matrix $A(x)$. Using this feature here gives,

$$
\mathbf{q}^{m}-\mathbf{q}^{l}=\alpha_{1}\left[\begin{array}{c}
-Z_{l} \\
1
\end{array}\right] \quad \text { and } \quad \mathbf{q}^{r}-\mathbf{q}^{m}=\alpha_{2}\left[\begin{array}{c}
Z_{r} \\
1
\end{array}\right]
$$

Combining the two equations yields

$$
\mathbf{q}^{r}-\mathbf{q}^{l}=\alpha_{1}\left[\begin{array}{c}
-Z_{l} \\
1
\end{array}\right]+\alpha_{2}\left[\begin{array}{c}
Z_{r} \\
1
\end{array}\right]=R_{l r}\left[\begin{array}{c}
\alpha_{1} \\
\alpha_{2}
\end{array}\right] .
$$

Thus one can obtain $\mathbf{q}^{m}(x, t)=\left[v^{m}(x, t), p^{m}(x, t)\right]^{T}$, after solving this linear system for $\alpha_{1}, \alpha_{2}$. When $-c_{l} t<x<c_{r} t$,

$$
\begin{aligned}
v^{m}(x, t) & =\frac{Z_{l} v_{L}+Z_{r} v_{R}}{Z_{l}+Z_{r}}-\frac{1}{Z_{l}+Z_{r}}\left(p_{R}-p_{L}\right), \\
p^{m}(x, t) & =\frac{Z_{r} p_{L}+Z_{l} p_{R}}{Z_{l}+Z_{r}}-\frac{Z_{l} Z_{r}}{Z_{l}+Z_{r}}\left(v_{R}-v_{L}\right) .
\end{aligned}
$$

Here the left-going and right-going waves are coupled together in the intermediate state $\mathbf{q}^{m}(x, t)$ though $R_{l r}$, which is composed by the left-going eigenvector from the left medium and the right-going eigenvector from the right medium.

In two or three dimensions, two-layer media is too simple to describe the structure of the real model. However, locally two-layer media structure is still a good approximation for a small piece of model where two different materials are presented and the interface is almost flat. The solution in 1D two-layer medium may extend to 2D or 3D over such a piece of model in a short period of travel time. In a long period of time, the extension does not work because the waves propagating from other places interfere with the solution in an unexpected way.

As seen in Fig.(2), the material parameters vary along the direction $\vec{n}\left(=\left(n_{x}, n_{y}\right)\right)$
at the interface, but keep constant along the tangent direction $\vec{\tau}$. After rotating the $x-y$ Cartesian system to $\xi-\eta$ coordinate system whose axes are along $\vec{n}$ and $\vec{\tau}$, the 2 D Riemann problem can be decoupled into two 1D Riemann problems with respect to $\xi$ and $\eta$. The 1D Riemann problem with respect to $\tau$ is trivial since material parameters and initial data are all homogeneous (constant) along the tangent direction. Along the normal direction, the acoustic wave equations in term of the pressure $p$ and normal velocity $v_{\vec{n}}\left(=n_{x} v_{x}+n_{y} v_{y}\right.$, where $v_{x}, v_{y}$ are velocity components) with piecewise constant initial data (Riemann problem) read,

$$
\begin{align*}
\rho(\xi) \frac{\partial v_{\vec{n}}}{\partial t}+\frac{\partial p}{\partial \xi}=0, & \frac{1}{\kappa(\xi)} \frac{\partial p}{\partial t}+\frac{\partial v_{\vec{n}}}{\partial \xi}=0  \tag{12}\\
\left(v_{\vec{n}}(\xi, 0), p(\xi, 0)\right) & = \begin{cases}\left(v_{\vec{n}}^{1}, p^{1}\right), & \xi<0 \\
\left(v_{\vec{n}}^{2}, p^{2}\right), & \xi>0\end{cases} \\
(\kappa(\xi), \rho(\xi)) & = \begin{cases}\left(\kappa_{1}, \rho_{1}\right), & \xi<0 \\
\left(\kappa_{2}, \rho_{2}\right), & \xi>0\end{cases}
\end{align*}
$$

where I use the fact that the field variables are independent of $\eta$ and thereby

$$
\begin{aligned}
\frac{\partial}{\partial x} & =\frac{\partial \xi}{\partial x} \frac{\partial}{\partial \xi}+\frac{\partial \eta}{\partial x} \frac{\partial}{\partial \eta}=n_{x} \frac{\partial}{\partial \xi}, \\
\frac{\partial}{\partial y} & =\frac{\partial \xi}{\partial y} \frac{\partial}{\partial \xi}+\frac{\partial \eta}{\partial y} \frac{\partial}{\partial \eta}=n_{y} \frac{\partial}{\partial \xi},
\end{aligned}
$$

For three dimensions, the Riemann problem in two-layer media for acoustics is decoupled into three 1D Riemann problem along the normal direction and two tangent directions of the interface plane. The only nontrivial Riemann problem again is the one along the normal direction, which is the same as Eqs.(12).

The Riemann solver discussed here is intended to construct numerical fluxes in


Figure 2: An illustration of two-layer media structure in 2D. $\vec{n}, \vec{\tau}$ are the unit normal vector and the unit tangent vector of the interface (dashed line), respectively. Here o is the origin.

DG methods, rather than analytic solutions. As mentioned before, two copies of degrees of freedom are defined at the same spatial point of the element edges in DG methods, since the DG solution has no continuity requirement. A reasonable solution at those points in a period of time combining information from both elements sharing the same edge is needed to advance the numerical solution. A Riemann problem in two-layer media fits for this setting ideally. For each point on an element edge, a small tube centered at this point and orthogonal to the edge is considered as the two-layer media model. The initial data are the numerical solutions at this common point from both elements. Since this point as time goes by stay in the middle region in the upper half $x-t$ plane as shown in Fig.(1), the intermediate state solution is the very ingredient to formulate the numerical flux.

The Riemann solver discussed above is fairly enough for the thesis. For other hyperbolic PDEs, especially when non-linearity presents, an exact Riemann solver is sometimes unlikely to build up. If needed, one may refer to Toro (1997), which discusses a variety of approximate Riemann solvers for fluid dynamics.

## Discontinuous Galerkin Time Domain Methods for Acoustics

In this section, I first derive a DG spatial discretization for Eqs.(1) . Then the time integration method used in this thesis is discussed. To complete the numerical method for acoustic wave equations, I also include the two types of boundary conditions for the numerical simulation.

## DG Spatial Discretization

The pressure-velocity formulation of the acoustic wave equations in 2D can be expressed as,

$$
\begin{align*}
\rho(x, z) \frac{\partial v_{x}}{\partial t}+\frac{\partial p}{\partial x} & =0 \\
\rho(x, z) \frac{\partial v_{z}}{\partial t}+\frac{\partial p}{\partial z} & =0  \tag{13}\\
\frac{1}{\kappa(x, z)} \frac{\partial p}{\partial t}+\frac{\partial v_{x}}{\partial x}+\frac{\partial v_{z}}{\partial z} & =w(t) \delta\left(x-x_{s}\right) \delta\left(z-z_{s}\right) .
\end{align*}
$$

To simplify the notation, I rewrite Eqs.(13) into a matrix-vector compact form,

$$
\begin{equation*}
\frac{\partial \mathbf{q}}{\partial t}+A(x, z) \frac{\partial \mathbf{q}}{\partial x}+B(x, z) \frac{\partial \mathbf{q}}{\partial z}=\left[0,0, \kappa w(t) \delta\left(x-x_{s}\right) \delta\left(z-z_{s}\right)\right]^{T} \tag{14}
\end{equation*}
$$

where $\mathbf{q}=\left[v_{x}, v_{y}, p\right]^{T}$,

$$
A(x, z)=\left[\begin{array}{ccc}
0 & 0 & 1 / \rho(x, z) \\
0 & 0 & 0 \\
\kappa(x, z) & 0 & 0
\end{array}\right] \quad B(x, z)=\left[\begin{array}{ccc}
0 & 0 & 0 \\
0 & 0 & 1 / \rho(x, z) \\
0 & \kappa(x, z) & 0
\end{array}\right]
$$

Denote $\mathscr{T}_{h}$ as a conforming triangulation of the computational domain $\bar{\Omega} \subset \mathbb{R}^{2}$ made of non-overlapping simplices $\mathcal{T}_{k}(k=1, \cdots, K)$, i.e., triangles, with radius of the inscribed circle denoted by $h_{k}$. The intersection of two triangles $\mathcal{T}_{j}$ and $\mathcal{T}_{k}$ is either
an edge or a vertex shared by them, or an empty set. For example, no neighbor triangles in the mesh are as shown in Fig.(3). In each triangle, the mass density and


Figure 3: fault intersection of two triangles, which is not allowed in a conforming triangulation
bulk modulus are assumed to be constant, that is $\rho\left(\left.\mathbf{x}\right|_{\mathcal{I}_{k}}=\rho_{k},\left.\kappa(\mathbf{x})\right|_{\mathcal{I}_{k}}=\kappa_{k}\right.$. Under this assumption the spatial derivatives of material parameters are excluded in the DG method.

Unlike finite element method giving a globally continuous approximation of the solutions, DG approximates the solutions on elements independently. On a triangular element $\mathcal{T}_{k}$, the interpolating Lagrange polynomials $\left\{l_{j}^{k}\right\}_{j}$ of degree $N\left(j=1, \cdots, N_{p}\right.$, where $\left.N_{p}=(N+1)(N+2) / 2\right)$ on the $\alpha$-optimized warp \& blend nodal set $\left\{\mathbf{x}_{j}^{k}\right\}_{j} \subset \overline{\mathcal{T}}_{k}$ $\left(l_{i}^{k}\left(\mathbf{x}_{j}^{k}\right)=\delta_{i j}\right)$ as indicated in Fig.(4) by Warburton (2006) are the basis functions for the spatial discretization. The numerical solution $\mathbf{q}_{h}$ of Eqs.(13) on $\mathcal{T}_{k}$ is then expressed as a linear combination of the polynomial basis functions as,

$$
\begin{align*}
\left.\mathbf{q}_{h}(\mathbf{x}, t)\right|_{\mathcal{I}_{k}} & =\left.\left[v_{x, h}(\mathbf{x}, t), v_{y, h}(\mathbf{x}, t), p_{h}(\mathbf{x}, t)\right]^{T}\right|_{\mathbf{T}_{k}} \\
& =\mathbf{q}_{h}^{k}(\mathbf{x}, t)=\left[v_{x, h}^{k}(\mathbf{x}, t), v_{z, h}^{k}(\mathbf{x}, t), p_{h}^{k}(\mathbf{x}, t)\right]^{T}=\sum_{j=1}^{N_{p}} \mathbf{q}_{h}\left(\mathbf{x}_{j}^{k}, t\right) l_{j}^{k}(\mathbf{x}) \\
& =\sum_{j=1}^{N_{p}} \mathbf{q}_{j}^{k}(t) l_{j}^{k}(\mathbf{x})=\sum_{j=1}^{N_{p}}\left[v_{x, j}^{k}(t), v_{z, j}^{k}(t), p_{j}^{k}(t)\right]^{T} l_{j}^{k}(\mathbf{x}) \tag{15}
\end{align*}
$$

The test function space $V_{h}$ is then defined as, $V_{h}=\bigoplus_{k=1}^{K} \operatorname{span}\left\{l_{j}^{k}\right\}_{j}$. This construction here is a nodal version of DG discussed in Hesthaven and Warburton (2008). The representation of $\mathbf{q}_{h}$ with Lagrange basis functions allows me to get the facial values at the boundary interpolating points directly. The advantage of using nodal basis function will be seen when computing the boundary integral and the numerical flux.


Figure 4: Illustrations of $\alpha$-optimized warp \& blend nodal distribution on the equilateral triangle. I use the $\alpha$-optimized warp \& blend nodes as the interpolating points of Lagrange polynomial basis functions. $N+1$ points live on each edge of the triangle when using basis functions of degree $N$. The three figures show the nodal distribution for different $N$. On the left, $N=1, N_{p}=3$; in the middle, $N=5, N_{p}=21$; on the right, $N=8, N_{p}=45$.

Multiplying Eqs.(14) by a test function $v_{h}$ in $V_{h}$ and taking integration over $\mathcal{T}_{k}$ yields,

$$
\int_{\mathcal{T}_{k}} v_{h} \frac{\partial \mathbf{q}}{\partial t}+\int_{\mathcal{T}_{k}} v_{h}\left(A \frac{\partial \mathbf{q}}{\partial x}+B \frac{\partial \mathbf{q}}{\partial z}\right) d V=\left[0,0, \kappa_{k} w(t) v_{h}\left(x_{s}, z_{s}\right)\right]^{T}
$$

Applying integration by parts gives,

$$
\int_{\mathcal{T}_{k}} v_{h} \frac{\partial \mathbf{q}}{\partial t}+\int_{\partial \mathcal{T}_{k}} v_{h} \mathbf{F} d S-\int_{\mathcal{T}_{k}}\left(\frac{\partial v_{h}}{\partial x} A \mathbf{q}+\frac{\partial v_{h}}{\partial z} B \mathbf{q}\right) d V=\left[0,0, \kappa_{k} w(t) v_{h}\left(x_{s}, z_{s}\right)\right]^{T}
$$

where $\mathbf{F}(\mathbf{x}, t)=\left(n_{x} A+n_{z} B\right) \mathbf{q}(\mathbf{x}, t)$ is the flux defined on $\mathcal{T}_{k}$ 's boundary $\partial \mathcal{T}_{k}$ of which $\vec{n}=\left[n_{x}, n_{z}\right]$ is the unit normal vector. Since I assume $A$ and $B$ are constant over $\mathcal{T}_{k}$, no spatial derivatives of $A$ and $B$ are shown in the above equation. Next, substituting
$q$ and $v_{h}$ with $\mathbf{q}_{h}^{k}$ and $l_{i}^{k}$ respectively and then partitioning $\partial \mathcal{T}_{k}$ into the edges $e_{m}^{k}$ ( $m=1,2,3$ ) shared by $\mathcal{T}_{k}$ 's neighbor triangle $\mathcal{T}_{k_{m}}$ gives the following called a DG scheme in weak form,

$$
\begin{align*}
\int_{\mathcal{T}_{k}} l_{i}^{k} \frac{\partial \mathbf{q}_{h}^{k}}{\partial t} & +\sum_{m=1}^{3} \int_{e_{m}^{k}} l_{i}^{k}\left(\mathbf{F}_{h}^{k, m}\right)^{*} d S \\
& -\int_{\mathcal{T}_{k}}\left(\frac{\partial l_{i}^{k}}{\partial x} A^{k} \mathbf{q}_{h}^{k}+\frac{\partial l_{i}^{k}}{\partial z} B^{k} \mathbf{q}_{h}^{k}\right) d V=\left[0,0, \kappa_{k} w(t) l_{i}^{k}\left(x_{s}, z_{s}\right)\right]^{T} \tag{16}
\end{align*}
$$

Several things need to be clarified here. Since the support of $l_{i}^{k}$ is $\overline{\mathcal{T}}_{k}, l_{i}^{k}\left(x_{s}, z_{s}\right)=0$ if $\left(x_{s}, y_{s}\right) \notin \overline{\mathcal{T}}_{k}$. For numerical stability consideration, I assume $\left(x_{s}, y_{s}\right) \in \stackrel{\circ}{\mathcal{T}}_{k}$ for some $k$. Therefore, the right hand side of Eq.(16) is nonzero only for one element. $\vec{n}^{k, m}=\left(n_{x}^{k, m}, n_{z}^{k, m}\right)$ is the unit normal vector of $e_{m}^{k}$ pointing from $\mathcal{T}_{k}$ to $\mathcal{T}_{k_{m}} . A^{k}$ and $B^{k}$ to denote the values of $A$ and $B$ on $\mathcal{T}_{k}$. On the left hand side of Eq. $(16),\left(\mathbf{F}_{h}^{k, m}\right)^{*}$ is called the numerical flux approximating $\mathbf{F}$ on $e_{m}^{k}$. to match with the approximation of the solution, $\left(\mathbf{F}_{h}^{k, m}\right)^{*}$ is defined as a linear combination of the traces of the basis functions along $e_{m}^{k}$. Because $l_{i}^{k}(\mathbf{x}) \equiv 0$ on $e_{m}^{k}$ if $x_{i}^{k} \notin e_{m}^{k}$ (since $\left.l_{i}^{k}(\mathbf{x})\right|_{e_{m}^{k}}$ is of degree $N$ and vanishes at $N+1$ points $),\left(\mathbf{F}_{h}^{k, m}\right)^{*}$ is the sum of $N+1$ terms associated with the $N+1$ interpolating points on $e_{m}^{k}$. Define $\mathbf{x}_{j}^{k, m}=\mathbf{x}_{n_{j}^{m}}^{k} \in e_{m}^{k}$ and $l_{j}^{k, m}=\left.l_{n_{j}^{m}}^{k}\right|_{e_{m}^{k}}$ $(j=1, \cdots, N+1)$ for some index $n_{j}^{m} .\left(\mathbf{F}_{h}^{k, m}\right)^{*}$ is then expressed as,

$$
\begin{align*}
\left(\mathbf{F}_{h}^{k, m}\right)^{*}(\mathbf{x}, t) & =\sum_{j=1}^{N+1}\left(n_{x}^{k, m} A^{k}+n_{z}^{k, m} B^{k}\right) \mathbf{q}_{h}^{*}\left(\mathbf{x}_{j}^{k, m}, t\right) l_{j}^{k, m}(\mathbf{x}) \\
& =\sum_{j=1}^{N+1}\left(n_{x}^{k, m} A^{k}+n_{z}^{k, m} B^{k}\right)\left(\mathbf{q}_{j}^{k, m}\right)^{*}(t) l_{j}^{k, m}(\mathbf{x})  \tag{17}\\
& =\sum_{j=1}^{N+1}\left(n_{x}^{k, m} A^{k}+n_{z}^{k, m} B^{k}\right)\left[\left(v_{x, j}^{k, m}\right)^{*}(t),\left(v_{y, j}^{k, m}\right)^{*}(t),\left(p_{j}^{k, m}\right)^{*}(t)\right]^{T} l_{j}^{k, m}(\mathbf{x})
\end{align*}
$$

in which $\mathbf{q}_{h}^{*}\left(\mathbf{x}_{j}^{k, m}, t\right)$ can be viewed as a reconstruction of the numerical solution $\mathbf{q}_{h}$ at $\mathbf{x}_{j}^{k, m}$. To see how $\mathbf{q}_{h}^{*}$ is related to the solution of a Riemann problem, expanding
$\left(n_{x}^{k, m} A+n_{z}^{k, m} B^{k}\right) \mathbf{q}_{h}^{*}\left(x_{j}^{k, m}, t\right)$ yields,

$$
\left[\frac{n_{x}^{k, m}}{\rho_{k}}\left(p_{j}^{k, m}\right)^{*}, \frac{n_{z}^{k, m}}{\rho_{k}}\left(p_{j}^{k, m}\right)^{*}, \kappa_{k}\left(n_{x}^{k, m}\left(v_{x, j}^{k, m}\right)^{*}+n_{z}^{k, m}\left(v_{z, j}^{k, m}\right)^{*}\right)\right]^{T} .
$$

One may notice that $\left[n_{x}^{k, m}\left(v_{x, j}^{k, m}+n_{z}^{k, m}\left(v_{z, j}^{k, m}\right)^{*}\right),\left(p_{j}^{k, m}\right)^{*}\right]^{T}:=\left[\left(v_{\vec{n}, j}^{k, m}\right)^{*},\left(p_{j}^{k, m}\right)^{*}\right]^{T}$ is the intermediate state solution of the local Riemann problem as stated in Eqs.(12) where the initial data on the left and right of the interface are $\mathbf{q}_{h}^{k}$ and $\mathbf{q}_{h}^{k_{m}}$ at $\mathbf{x}_{j}^{k, m}$. The benefit of using nodal basis functions is that evaluation of $\mathbf{q}_{h}^{k}, \mathbf{q}_{h}^{k_{m}}$ at $\mathbf{x}_{j}^{k, m}$ costs nothing, recalling that $\mathbf{x}_{j}^{k, m}$ is an alias of an interpolating point $\mathbf{x}_{n_{j}^{m}}^{k}$ of $\mathcal{T}_{k}$ for some index $n_{j}^{m}$ and by symmetry of the nodal points $\mathbf{x}_{j}^{k, m}$ is also an interpolating point of $\mathcal{T}_{k_{m}}$ on $e_{m}^{k}$.

Applying Green's formula once again in Eq.(16) gives the DG scheme in strong form for acoustic wave equations,

$$
\begin{align*}
\int_{\mathcal{T}_{k}} l_{i}^{k} \frac{\partial \mathbf{q}_{h}^{k}(t)}{\partial t} & +\sum_{m=1}^{3} \int_{e_{m}^{k}} l_{i}^{k}\left(\left(\mathbf{F}_{h}^{k, m}\right)^{*}-\mathbf{F}_{h}^{k, m}\right) d S \\
& +\int_{\mathcal{T}_{k}}\left(A^{k} \frac{\partial \mathbf{q}_{h}^{k}}{\partial x}+B^{k} \frac{\partial \mathbf{q}_{h}^{k}}{\partial z}\right) l_{i}^{k} d V=\left[0,0, \kappa_{k} w(t) l_{i}^{k}\left(x_{s}, z_{s}\right)\right]^{T} \tag{18}
\end{align*}
$$

where $\mathbf{F}_{h}^{k, m}=\left(n_{x}^{k, m} A+n_{z}^{k, m} B\right) \mathbf{q}_{h}^{k}$. Substituting $\mathbf{q}_{h}^{k}$ in Eq.(18) with the linear combination of basis functions as indicated in Eq.(15) completes the spatial DG discretization construction and yields a time dependent ordinary differential equation (ODE) system,

$$
\begin{align*}
\sum_{j=1}^{N_{p}} \frac{\partial \mathbf{q}_{j}^{k}(t)}{\partial t}\left(l_{i}^{k}, l_{j}^{k}\right)_{\mathcal{T}_{k}} & +\sum_{m=1}^{3} \sum_{j=1}^{N+1}\left(n_{x}^{k, m} A^{k}+n_{z}^{k, m} B^{k}\right)\left(\left(\mathbf{q}_{j}^{k, m}\right)^{*}(t)-\mathbf{q}_{n_{j}^{m}}^{k}(t)\right)\left(l_{i}^{k}, l_{j}^{k, m}\right)_{e_{m}^{k}} \\
& +\sum_{j=1}^{N_{p}} A^{k} \mathbf{q}_{j}^{k}(t)\left(l_{i}^{k}, \frac{\partial l_{j}^{k}}{\partial x}\right)_{\mathcal{T}_{k}}+\sum_{j=1}^{N_{p}} B^{k} \mathbf{q}_{j}^{k}(t)\left(l_{i}^{k}, \frac{\partial l_{j}^{k}}{\partial z}\right)_{\mathcal{T}_{k}}  \tag{19}\\
& =\left[0,0, \kappa_{k} w(t) l_{i}^{k}\left(x_{s}, y_{s}\right)\right]^{T}
\end{align*}
$$

where $(\cdot, \cdot)_{\mathcal{T}_{k}},(\cdot, \cdot)_{e_{m}^{k}}$ are the standard $L^{2}$ inner product over $\mathcal{T}_{k}$ and $e_{m}^{k}$. Gathering Eq.(19) for each $l_{i}^{k}$ together then gives a compact matrix-vector form by introducing some new notations. First denote,

$$
\begin{aligned}
& \mathbf{v}_{x}^{k}(t)=\left[v_{x, 1}^{k}(, t), \cdots, v_{x, N_{p}}^{k}(t)\right]^{T}, \\
& \mathbf{v}_{z}^{k}(t)=\left[v_{z, 1}^{k}(t), \cdots, v_{z, N_{p}}^{k}(t)\right]^{T}, \\
& \mathbf{p}^{k}(t)=\left[p_{1}^{k}(t), \cdots, p_{N_{p}}^{k}(t)\right]^{T}, \\
& \left(\mathbf{p}^{k, m}\right)^{*}(t)-\mathbf{p}^{k, m}(t)=\left[\left(p_{1}^{k, m}\right)^{*}(t)-p_{n_{1}^{m}}^{k}(t), \cdots,\left(p_{N+1}^{k, m}\right)^{*}(t)-p_{n_{N+1}^{m}}^{k}(t)\right]^{T}, \\
& \left(\mathbf{v}_{\vec{n}}^{k, m}\right)^{*}(t)=\left[\left(v_{\vec{n}, 1}^{k, m}\right)^{*}(t), \cdots,\left(v_{\vec{n}, N+1}^{k, m}\right)^{*}(t)\right]^{T}, \\
& \mathbf{v}_{\vec{n}}^{k, m}(t)=\left[v_{\vec{n}, n_{1}^{m}}^{k, m}(t), \cdots, v_{\vec{n}, n_{N+1}^{m}}^{k, m}(t)\right]^{T},
\end{aligned}
$$

where $v_{\vec{n}, n_{j}^{m}}^{k, m}=n_{x}^{k, m} v_{x, n_{j}^{m}}^{k, m}+n_{y}^{k, m} v_{z, n_{j}^{m}}^{k, m}$. Then define the local volume mass matrix $M^{k} \in \mathbb{R}^{N_{p} \times N_{p}}$, the local stiffness matrices $S^{x, k}, S^{z, k} \in \mathbb{R}^{N_{p} \times N_{p}}$, the local edge mass $\operatorname{matrix} M^{k, m} \mathbb{R}^{N_{p} \times(N+1)}$ as,

$$
M_{i j}^{k}=\left(l_{i}^{k}, l_{j}^{k}\right)_{\mathcal{I}_{k}}, S_{i j}^{x, k}=\left(l_{i}^{k}, \frac{\partial l_{j}^{k}}{\partial x}\right)_{\mathcal{I}_{k}}, S_{i j}^{z, k}=\left(l_{i}^{k}, \frac{\partial l_{j}^{k}}{\partial z}\right)_{\mathcal{I}_{k}}, M_{i j}^{k, m}=\left(l_{i}^{k}, l_{j}^{k, m}\right)_{e_{m}^{k}} .
$$

With these notations the compact matrix-vector form of Eq.(19) can be expressed as,

$$
\begin{aligned}
M^{k} \frac{d \mathbf{v}_{x}^{k}(t)}{d t}+\frac{1}{\rho_{k}} S^{x, k} \mathbf{p}^{k}(t)+\sum_{m=1}^{3} \frac{n_{x}^{e_{m}^{k}}}{\rho_{k}} M^{k, e_{m}^{k}}\left(\left(\mathbf{p}^{k, m}\right)^{*}(t)-\mathbf{p}^{k, m}(t)\right)=0, \\
M^{k} \frac{d \mathbf{v}_{z}^{k}(t)}{d t}+\frac{1}{\rho_{k}} S^{z, k} \mathbf{p}^{k}(t)+\sum_{m=1}^{3} \frac{n_{z}^{e_{m}^{k}}}{\rho_{k}} M^{k, e_{m}^{k}}\left(\left(\mathbf{p}^{k, m}\right)^{*}(t)-\mathbf{p}^{k, m}(t)\right)=0, \\
M^{k} \frac{d \mathbf{p}^{k}(t)}{d t}+\kappa_{k} S^{x, k} \mathbf{v}_{x}^{k}(t)+\kappa_{k} S^{z, k} \mathbf{v}_{z}^{k}(t)+\sum_{m=1}^{3} \kappa_{k} M^{k, e_{m}^{k}}\left(\left(\mathbf{v}_{\vec{n}}^{k, m}\right)^{*}(t)-\mathbf{v}_{\vec{n}}^{k, m}(t)\right)=\kappa_{k} w(t) \mathbf{l}^{k}\left(\mathbf{x}_{s}\right),
\end{aligned}
$$

where $\mathbf{l}^{k}\left(\mathbf{x}_{s}\right)=\left[l_{1}^{k}\left(\mathbf{x}_{s}\right), \cdots, l_{N_{p}}^{k}\left(\mathbf{x}_{s}\right)\right]^{T}$. Furthermore multiplying the above equations
by $\left(M^{k}\right)^{-1}$ yields an fully explicit form,

$$
\begin{align*}
\frac{d \mathbf{v}_{x}^{k}(t)}{d t} & =-\frac{1}{\rho_{k}} D^{x, k} \mathbf{p}^{k}(t)-\sum_{m=1}^{3} \frac{n_{x}^{e_{m}^{k}}}{\rho_{k}} L^{k, m}\left(\left(\mathbf{p}^{k, m}\right)^{*}(t)-\mathbf{p}^{k, m}(t)\right), \\
\frac{d \mathbf{v}_{z}^{k}(t)}{d t} & =-\frac{1}{\rho_{k}} D^{z, k} \mathbf{p}^{k}(t)-\sum_{m=1}^{3} \frac{n_{y}^{e_{m}^{k}}}{\rho_{k}} L^{k, m}\left(\left(\mathbf{p}^{k, m}\right)^{*}(t)-\mathbf{p}^{k, m}(t)\right),  \tag{20}\\
\frac{d \mathbf{p}^{k}(t)}{d t} & =-\kappa_{k} D^{x, k} \mathbf{v}_{x}^{k}(t)-\kappa_{k} D^{z, k} \mathbf{v}_{z}^{k}(t) \\
& -\sum_{m=1}^{3} \kappa_{k} L^{k, m}\left(\left(\mathbf{v}_{\vec{n}}^{k, m}\right)^{*}(t)-\mathbf{v}_{\vec{n}}^{k, m}(t)\right)+\kappa_{k} w(t)\left(M^{k}\right)^{-1} \mathbf{l}^{k}\left(\mathbf{x}_{s}\right),
\end{align*}
$$

where $D^{x, k}=\left(M^{k}\right)^{-1} S^{x, k}, D^{z, k}=\left(M^{k}\right)^{-1} S^{z, k}, L^{k, m}=\left(M^{k}\right)^{-1} M^{k, e_{m}^{k}}$. To reduce the storage, each element $\mathcal{T}_{k}$ is considered to be the image of the reference triangle $\hat{\mathcal{D}}=\{(r, s) \mid-1 \leq r, s ; r+s \leq 0\}$. Then for example the mass matrix $M^{k}$ for the $\mathrm{k}^{\prime}$ 'th element is given by

$$
\begin{equation*}
M_{i j}^{k}=\int_{\hat{\mathcal{D}}} l_{i}(r, s) l_{j}(r, s) J^{k}(r, s) d r d s \tag{21}
\end{equation*}
$$

where $\left\{l_{j}(r, s)\right\}_{j=1}^{N_{p}}$ are the basis functions on $\hat{\mathcal{D}}$. Since I use straight sided element now, the Jacobian $J^{k}$ is a constant, independent of $(r, s)$. The mass matrix for $\mathcal{T}_{k}$ can be expressed as a scalar $J^{k}$ multiple of the mass matrix on the reference element $\hat{\mathcal{D}}$,

$$
\begin{equation*}
M_{i j}^{k}=J^{k} M_{i j} \tag{22}
\end{equation*}
$$

where $M_{i j}=\int_{\hat{\mathcal{D}}} l_{i}(r, s) l_{j}(r, s) d r d s$. We only need to store the mass matrix on the reference element or its inverse. In the same way the stiffness matrices and the edge mass matrices is stored only for the reference element. For the edge mass matrices, the reference element becomes $I=\{r \mid-1 \leq r \leq 1\}$. To get the explicit formulation of these matrices, one may refer to Hesthaven and Warburton (2008).

Now it can be seen clearly that once the numerical fluxes are computed, the time
update is explicit and completely independent for each element. This feature is useful especially for parallelization.

## Time Discretization

After the spatial discretization, an ODE system needs to be solved to march the numerical solution along time,

$$
\begin{equation*}
\frac{d \mathbf{Q}_{h}^{k}}{d t}=\mathcal{R}_{h}\left(\mathbf{Q}_{h}^{k},\left(\mathbf{Q}_{h}^{k, 1}\right)^{*},\left(\mathbf{Q}_{h}^{k, 2}\right)^{*},\left(\mathbf{Q}_{h}^{k, 3}\right)^{*}, t\right) \tag{23}
\end{equation*}
$$

where $\mathcal{R}_{h}$ is the operator grouping the right hand side terms in Eq.(20) together, and

$$
\mathbf{Q}_{h}^{k}=\left[\begin{array}{c}
v_{x}^{k} \\
v_{y}^{k} \\
\left(p^{k}\right)
\end{array}\right], \quad\left(\mathbf{Q}_{h}^{k, m}\right)^{*}=\left[\begin{array}{c}
\left(v_{n}^{k, m}\right)^{*} \\
\left(p^{k, m}\right)^{*}
\end{array}\right]
$$

In this thesis, I use the low-storage five-stage fourth-order explicit Runge-Kutta (RK) method to solve Eq.(23) by Carpenter and Kennedy (1994),

$$
\begin{aligned}
& \text { for each } \mathrm{k} \\
& \mathbf{U}_{0}^{k}=\mathbf{Q}_{h}^{k}(t), \mathbf{h}_{0}^{k}=0 \\
& \text { end } \\
& \text { for } \mathrm{i}=1: 5 \\
& \text { for each k } \\
& \quad \text { update }\left(\mathbf{Q}_{h}^{k, m}\right)^{*} \text { with } \mathbf{U}_{i}^{k} \text { and } \mathbf{U}_{i}^{k_{m}}, m=1,2,3 \\
& \quad \mathbf{h}_{i}^{k}=a_{i} \mathbf{h}_{i-1}^{k}+\Delta t \mathcal{R}_{h}\left(\mathbf{U}_{i-1}^{k},\left(\mathbf{Q}_{h}^{k, 1}\right)^{*},\left(\mathbf{Q}_{h}^{k, 2}\right)^{*},\left(\mathbf{Q}_{h}^{k, 3}\right)^{*}, t+c_{i} \Delta t\right) \text {, } \\
& \quad \mathbf{U}_{i}^{k}=\mathbf{U}_{i-1}^{k}+b_{i} \mathbf{h}_{i} \\
& \text { end } \\
& \text { end } \\
& \mathbf{Q}_{h}^{k}(t+\Delta t)=\mathbf{U}_{5}^{k} \text {. }
\end{aligned}
$$

The coefficients $a_{i}, b_{i}, c_{i}$ can be found in Tab.(1). The advantage of using such scheme is the low memory usage, since only one additional storage is required. One more stage is present in this fourth-order scheme and seems to add cost. However, this scheme allows to use a larger time step $\Delta t$ and thus reduce the overall computation cost.

For numerical stability, the time step $\Delta t$ should not be taken too large. As proved, the high-order accurate RK DG methods require $\Delta t \sim \mathcal{O}(\Delta x)$. To be specific, the time step I use here has a bound like,

$$
\begin{equation*}
\Delta t \leq C \frac{\min _{k} h_{k}}{(N+2)(N+1)} \tag{24}
\end{equation*}
$$

where $h_{k}$ indicates the radius of the inscribed circle in the element $\mathcal{T}_{k}$.

Table 1: Coefficients for the low-storage five-stage fourth-order explicit RK method

| $i$ | $a_{i}$ | $b_{i}$ | $c_{i}$ |
| :---: | :---: | :---: | :---: |
| 1 | 0 | $\frac{1432997174477}{9575080441755}$ | 0 |
| 2 | $-\frac{567301805773}{1357537059087}$ | $\frac{5161836677717}{13612068292357}$ | $\frac{1432997174477}{9575080441755}$ |
| 3 | $-\frac{2404267990393}{2016746695238}$ | $\frac{1720146321549}{2090206949498}$ | $\frac{2526269341429}{6820363962896}$ |
| 4 | $-\frac{3550918686646}{2091501179385}$ | $\frac{3134564353537}{4481467310338}$ | $\frac{2006345519317}{3224310063776}$ |
| 5 | $-\frac{1275806237668}{842570457699}$ | $\frac{2277821191437}{14882151754819}$ | $\frac{2802321613138}{2924317926251}$ |

## Boundary Conditions

In this thesis, numerical simulation is always carried out on a bounded domain, whose boundaries mimics either the physical sea surface, landform or the fields far away from the domain of interest. The free surface boundary condition and absorbing boundary condition are two types of boundary conditions exclusively used in the seismic simulation.

Due to the high contrast discontinuities of material parameters between water and air, soil and air, waves are almost reflected back with a neglected amount of energy passing through when they hit the surface. The reflection boundary condition is used to replicate this action of waves in the numerical simulation. Till now, I always assume that $\mathcal{T}_{k}$ has three neighbor elements $\mathcal{T}_{k_{m}}$. But when the $\mathcal{T}_{k}$ 's edge $e_{m}^{k}$ belongs to the free surface boundary (the sea surface or landforms), $\mathcal{T}_{k_{m}}$ does not exist in the triangulation $\mathscr{T}_{h}$. I have to assign the initial data on the right of the interface in
the Riemann problem Eqs.(12) so as to compute the numerical flux $\left(F_{h}^{k, m}\right)^{*}$ on $e_{m}^{k}$. Applying the free surface boundary condition to the local Riemann problem yields the initial data of Eqs.(12) as

$$
v_{\vec{n}}^{2}=-v_{\vec{n}}^{1}, \quad p^{2}=p^{1} .
$$

This is an analogy to the free surface boundary condition in finite difference method in which the free surface boundary condition is applied to ghost grids in order to update the numerical solutions defined on the computational grids whose stencil is out of the computational domain.

In the seismic simulation, the domain of interest is relatively small compared to the distance waves can propagate. Absorbing boundary condition does not try to mimic any physical scenarios, but is used to truncate the open domain problem into a finite one so that the numerical method can handle. In this thesis, I use a perfectly matched layer (PML), one kind of the absorbing boundary condition first designed by Berenger (1994) for Maxwell's equations. PML can be considered as layers wrapping the original computational domain to absorb outgoing incident wave from any incidence angle without false reflection. A PML version of acoustic wave
equations proposed by Abarbanel and Gottlieb (1998) can be stated as

$$
\begin{align*}
\frac{\partial v_{x}}{\partial t}+2 \eta_{x} v_{x}+\frac{1}{\rho(x, z)} \frac{\partial p}{\partial x} & =\eta_{x} P_{x} \\
\frac{\partial v_{z}}{\partial t}+2 \eta_{z} v_{z}+\frac{1}{\rho(x, z)} \frac{\partial p}{\partial z} & =\eta_{z} P_{z} \\
\frac{\partial p}{\partial t}+\kappa(x, z) \nabla \cdot \mathbf{v}+\eta_{x_{p}} Q_{x}+\eta_{z_{p}} Q_{z} & =\kappa(x, z) w(t) \delta\left(\mathbf{x}-\mathbf{x}_{s}\right),  \tag{25}\\
\frac{\partial P_{x}}{\partial t}+\eta_{x} v_{x} & =0 \\
\frac{\partial P_{y}}{\partial t}+\eta_{y} v_{z} & =0 \\
\frac{\partial Q_{x}}{\partial t}+\eta_{x} Q_{x} & =\kappa(x, z) v_{x} \\
\frac{\partial Q_{z}}{\partial t}+\eta_{z} Q_{z} & =\kappa(x, z) v_{z}
\end{align*}
$$

where $\left(P_{x}, P_{z}, Q_{x}, Q_{z}\right)$ are four auxiliary variables, and

$$
\eta_{\alpha}=\left\{\begin{array}{ll}
\eta_{\alpha_{\max }}\left(\frac{L_{\alpha} / 2+\alpha}{d}\right)^{2} & \alpha \in\left[-d-L_{\alpha} / 2,-L_{\alpha} / 2\right] \\
0 & \alpha \in\left(-L_{\alpha} / 2, L_{\alpha} / 2\right] \\
\eta_{\alpha_{\max }}\left(\frac{L_{\alpha} / 2-\alpha}{d}\right)^{2} & \alpha \in\left(L_{\alpha} / 2, L_{\alpha} / 2+d\right]
\end{array} \quad \eta_{\alpha_{p}}=\frac{d \eta_{\alpha}}{d \alpha}\right.
$$

where $\alpha \in\{x, z\}$, and $\left\{L_{x}, L_{z}\right\}$ is the domain size, and $d$ is the PML thickness. One can see that Eq.(25) inside the domain is exactly the original acoustic wave equations and in the PML the auxiliary variables work as damping sources to absorb the acoustic pressure as well as the velocity. Further more, the auxiliary variables are governed by ordinary differential equations, which can be computed with very little cost.

## Finite Difference Time Domain Method for Acoustics

In this section, I present the widely used 2-4 (second-order in time and fourthorder in space) staggered-grid finite difference method for Eqs.(13). This method
later is used to compare with DGTD methods described before.

Two-dimensional staggered grids for Eqs.(13) employ three different sets of grids on which the field variables $p, v_{x}$ and $v_{z}$ are computed as shown in Fig.(5). The 2-4


Figure 5: 2D staggered grid for the pressure-velocity formulation of acoustic wave equations. $h_{x}$ and $h_{z}$ denote spatial steps along $x$-axis and $z$-axis. Pressure grids are represented by circles. Horizontal and vertical velocity grids are represented by squares and triangles respectively.
staggered-grid finite different method for Eqs.(13) is given as follows,

$$
\begin{aligned}
p_{i, j}^{n} & =p_{i, j}^{n-1}-\Delta t \frac{\kappa_{i, j}}{h_{x}}\left(-\frac{1}{24}\left(v_{x}\right)_{i+3 / 2, j}^{n-1 / 2}+\frac{9}{8}\left(v_{x}\right)_{i+1 / 2, j}^{n-1 / 2}-\frac{9}{8}\left(v_{x}\right)_{i-1 / 2, j}^{n-1 / 2}+\frac{1}{24}\left(v_{x}\right)_{i-3 / 2, j}^{n-1 / 2}\right) \\
& -\Delta t \frac{\kappa_{i, j}}{h_{z}}\left(-\frac{1}{24}\left(v_{z}\right)_{i, j+3 / 2}^{n-1 / 2}+\frac{9}{8}\left(v_{z}\right)_{i, j+1 / 2}^{n-1 / 2}-\frac{9}{8}\left(v_{z}\right)_{i, j-1 / 2}^{n-1 / 2}+\frac{1}{24}\left(v_{z}\right)_{i, j-3 / 2}^{n-1 / 2}\right) \\
& +\Delta t \kappa_{i, j} w^{n-1 / 2} \delta_{i, i_{s}} \delta_{j, j_{s}}, \\
\left(v_{x}\right)_{i+1 / 2, j}^{n+1 / 2} & =\left(v_{x}\right)_{i+1 / 2, j}^{n-1 / 2}-\Delta t \frac{1}{\rho_{i+1 / 2, j} h_{x}}\left(-\frac{1}{24} p_{i+2, j}^{n}+\frac{9}{8} p_{i+1, j}^{n}-\frac{9}{8} p_{i, j}^{n}+\frac{1}{24} p_{i-1, j}^{n}\right), \\
\left(v_{z}\right)_{i, j+1 / 2}^{n+1 / 2} & =\left(v_{z}\right)_{i, j+1 / 2}^{n-1 / 2}-\Delta t \frac{1}{\rho_{i, j+1 / 2} h_{z}}\left(-\frac{1}{24} p_{i, j+2}^{n}+\frac{9}{8} p_{i, j+1}^{n}-\frac{9}{8} p_{i, j}^{n}+\frac{1}{24} p_{i, j-1}^{n}\right),
\end{aligned}
$$

where $\Delta t$ is the time step, $p_{i, j}^{n}$ for example denotes the value of the pressure variable at $\left(i h_{x}, j h_{z}\right)$ and time $n \Delta t$, and $i_{s}, j_{s}$ are the indices such that $x_{s}=i_{s} h_{x}$ and $z_{s}=j_{s} h_{z}$.

In this thesis, I use an open source software IWAVE by Terentyev (2008) as the FD solver. IWAVE provides a general FD framework for solving time domain PDE on both distributed and shared memory computer architectures. In IWAVE, one can design FD schemes by providing the associated stencil information. The 2-4 staggered-grid finite difference method I use for comparison has been implemented in IWAVE. This saves me a lot of time to implement my own version of this FD scheme so that I can focus myself on the DG implementation and the comparison.

## RESULTS CHAPTER

## Introduction

In this section, I present numerical examples regarding the convergence tests of DG methods and the numerical comparison between DGTD methods and FDTD methods. First, I make convergence tests of DGTD methods for two cases: point source wave in the homogeneous medium and the plane wave propagating in two-layer medium. Since I have the analytic solutions for both cases, I can estimate the error and convergence rate precisely, and compare these results with the ones in theory. The purpose of these tests is to make sure that my DGTD implementation is valid. As mentioned before, interface error can downgrade the convergence rate of staggered-grid FD methods to 1st order. This error also appears in DGTD methods. In my DGTD implementation, each element in the triangulation is considered as a homogeneous media. If the mesh misaligns with the material interface, the approximate model on this mesh would falsely represent the real model and then produce the interface error. Through numerical examples, I display the interface error in DGTD methods. Examples of
two-layer media on the interface conforming mesh and the local refined mesh exhibit two ways to reduce this error in some cases. At last, the comparison of DGTD and FDTD is carried out on two examples. Since analytic solutions are inaccessible in both examples, I first discuss the error estimate via Richardson extrapolation. Then I compare the computing cost measured in GFLOP (approximately $10^{9}$ float point operations) for the two solutions by DGTD and FDTD to have the roughly the same accuracy (5\%).

## Convergence tests of DGTD

I make convergence tests of DGTD methods through point source wave and the plane wave. In the point source wave case, I measure the error of a trace at a given point on a series of globally refined meshes. In the plane wave case, the $L^{2}$ error of the pressure at a given time is computed to estimate the convergence rate. In both cases, DGTD methods behave the way as theories predict.

## Point Source Wave

The analytic solution for a point source wave has been given as a spherical expanding wave in Eq.(4). The computation domain $[-0.5,0.5] \times[-0.5,0.5]$ is a unit square centered at the origin. The mirror reflection boundary condition is applied. The material is homogeneous with $\rho=1.0, c=1.0$. A point source is located at $\mathbf{x}_{s}=(0,1 / 4)$ with the source pulse as,

$$
\begin{equation*}
w(t)=\left(t-t_{0}\right) e^{-\left(\pi f_{0}\left(t-t_{0}\right)\right)^{2}}, \tag{26}
\end{equation*}
$$

where $f_{0}=10, t_{0}=1.2 / f_{0}$. A trace is recorded at $\mathbf{x}_{r}=(0,-1 / 4)$. The basis functions used are of degree 5. This example is carried out on a series of globally refined meshes
with grid size $1 / 10,1 / 20,1 / 40,1 / 80$, respectively. As shown in Fig.(6), the trace error decreases $10^{2}$ times $\left(\approx 2^{6}\right.$, approximately) till the single floating point precision, when the grid size decreases by half.


Figure 6: Trace error for the point source wave example.

## Plane Wave

In this example, a plane wave propagates in a two-layer media with $90^{\circ}$ incident angle at the material interface. Though essentially this is a 1D problem, it provides a good example in 2 D to test my DGTD implementation. The material interface is assumed to be a vertical line at $x=0$. I use $\left(\rho_{l}, c_{l}\right)$ and $\left(\rho_{r}, c_{r}\right)$ to denote the density and the wave speed for the materials at the two sides of the interface,

$$
\begin{aligned}
& \rho_{l}=2100 \mathrm{~kg} / \mathrm{m}^{3}, \quad c_{l}=2.3 \mathrm{~m} / \mathrm{ms} \\
& \rho_{r}=2300 \mathrm{~kg} / \mathrm{m}^{3}, \quad c_{r}=3.0 \mathrm{~m} / \mathrm{ms}
\end{aligned}
$$

If the initial particle velocity at $y$ direction $v_{y}$ is 0 and other initial field variables $\left(v_{x}, p\right)$ are smooth, then Eqs.(7) give me the analytic solutions for this case. Here I pick $g$ as a Ricker's wavelet with central frequency $f_{0}=10 \mathrm{~Hz}$,

$$
g(t)=\left(1-2\left(\pi f_{0}\left(t-t_{0}\right)\right)^{2}\right) e^{-\left(\pi f_{0}\left(t-t_{0}\right)\right)^{2}}
$$

The computation domain $\Omega$ is $[0,1800 \mathrm{~m}] \times[-15 \mathrm{~m}, 15 \mathrm{~m}]$ and the simulation time is 600 ms . I do the convergence test on a series of globally refined interface-fitting meshes as indicated in Fig.(7) and basis functions with different order. The estimated convergence rate shown in Tab.(2) agree to the optimal convergence rate proven in Lesaint and Raviart (1974); Johnson and Pitkaranta (1986).


Figure 7: Illustration of the interface-fitting mesh. Blue and green stand for two material.

Table 2: Convergence test for the plane wave case on the interface conforming meshes. $N$ indicates the degree of basis functions in the DGTD method. The $L^{2}$ errors at $T=600 \mathrm{~ms}$ are measured for field variables. $R$ denotes the estimated convergence rate based on the $L^{2}$ error of the pressure.

| $h$ | $N$ | $\left\\|p_{h}(\cdot, T)-p(\cdot, T)\right\\|_{L^{2}}$ | $\left\\|u_{h}(\cdot, T)-u(\cdot, T)\right\\|_{L^{2}}$ | $\left\\|v_{h}(\cdot, T)-v(\cdot, T)\right\\|_{L^{2}}$ | R |
| :---: | :---: | :---: | :---: | :---: | :---: |
| 10 | 1 | 0.7649 | 0.7762 | 0.1194 | 2.86 |
| 5 | 1 | 0.1053 | 0.1102 | 0.0307 | 2.74 |
| 2.5 | 1 | 0.0157 | 0.0177 | 0.0077 | - |
| 10 | 2 | 0.0084 | 0.0098 | 0.0044 | 3 |
| 5 | 2 | 0.0010 | 0.0012 | $5.54 \mathrm{e}-4$ | 2.95 |
| 2.5 | 2 | $1.29 \mathrm{e}-4$ | $1.50 \mathrm{e}-5$ | $6.90 \mathrm{e}-5$ | - |

## Interface Error

In this section, I illustrate the interface error in DGTD method by the plane wave example when the triangular mesh misaligns with the material interface. One can see that the interface error is associated with the false representation of the model and therefore can not be eliminated by higher order schemes. Without modifying the DGTD method, I propose two mesh techniques: interface conforming mesh and local mesh refinement.

I use the same model and parameters as the plane wave example. The trace recorded at [500 m, 0] as shown in Fig.(10) has two spikes corresponding to the direct wave and the reflected wave. The error in the direct wave is associated with the truncation error of the scheme and therefore can be reduced by higher order schemes, while the error in the reflected wave is associated with the interface error. I use three sets of triangular meshes to test this example: the interface conforming mesh, the mesh misaligned with the interface and the local refined mesh near the interface as shown in Fig.(7), Fig.(8) and Fig.(9), respectively. Fig.(10) shows the traces of the


Figure 8: Illustration of the mesh misaligned with the interface.
analytic and numerical solution recorded at [500 m,0] computed by DGTD with basis functions of degree 1 and 2 on the interface conforming mesh. Both the direct wave and the reflected wave are resolved better by a high order scheme (basis functions of degree 2). Fig.(11) shows the trace of the analytic and numerical solution at the same receiver computed by DGTD with basis functions of degree 1,2 and 4 on the


Figure 9: Illustration of the local refined mesh near the interface.
misaligned mesh. The direct wave is resolved accurately by high order scheme. But the first-order error caused by the mesh misalignment dominates the error in the reflected wave. Fig.(12) illustrates the same simulation but on the local refined mesh. The time-shift effect in the reflected wave is much smaller than on the misaligned mesh.

## Comparison of DGTD and FDTD

In the seismic community, people more care about the trace (seismogram) error at the near surface receivers (geophones) rather than the error measured over the entire computation domain. Hence the trace error for each receiver is measured for the following numerical experiments for the purpose of comparison involving DGTD and FDTD. Due to lack of analytic solutions for most of realistic models, the numerical error is estimated by Richardson extrapolation as usual, that is, if assuming the numerical solution $D(h)$ differs from the analytic solution $\bar{D}$ by $E(h)=C h^{R}+$ $O\left(h^{R+1}\right)$, then

$$
\begin{equation*}
E(h) \simeq \frac{D(2 h)-D(h)}{2^{R}-1} \tag{27}
\end{equation*}
$$



Figure 10: Traces of the analytic and numerical solutions recorded at [500 m,0] on the interface conforming mesh as shown in Fig.(7). Basis functions of degree 1 and 2 are used to compute the trace plotted on the left and on the right, respectively. The top plots are the entire traces from 0 ms to 600 ms . The middle plots show the direct wave corresponding to the first spike of the entire trace. The bottom plots show the reflected wave corresponding to the second spike of the entire trace.


Figure 11: Traces of the analytic and numerical solutions recorded at [500 m,0] on the misaligned mesh as shown in Fig.(8). Basis functions of degree 2 and 4 are used to compute the traces plotted on the left and right, respectively. The plots at the top show the direct wave, while the ones at the bottom show the reflected wave.


Figure 12: Traces of the analytic and numerical solutions recorded at [500 m,0] on the local refined mesh as shown in Fig.(9) . Basis functions of degree 1 and 2 are used to compute the traces plotted on the left and right, respectively. The plots at the top show the direct wave, while the ones at the bottom show the reflected wave.
where $R$ can be estimated by having $E(2 h)$,

$$
\begin{equation*}
R \simeq \log _{2} \frac{E(2 h)}{E(h)} \tag{28}
\end{equation*}
$$

The programs for FDTD and DGTD methods are written in the ISO C language and use MPI and domain decomposition for parallelization. The numerical experiments were performed in single precision on a 2.66 GHz Intel Core2 Quad Q9450 CPU. The operating system is Linux (2.6.18 kernel) and the compiler is GNU C compiler (version 4.1.2). For both examples, the numerical cost is mearsured by the total number of floating point operations (GFLOP) and wall clock time.

## Square-circle model

In this example, the computation domain is a $[-500 \mathrm{~m}, 500 \mathrm{~m}] \times[-500 \mathrm{~m}, 500 \mathrm{~m}]]$ square with two different media separated by a circle of radius 125 m at the center. As shown in Fig.(13), the parameters for the simulation are defined as,

- inside the circle: $\rho=1000 \mathrm{~kg} / \mathrm{m}^{3}, c=1000 \mathrm{~m} / \mathrm{s}$;
- outside the circle: $\rho=1500 \mathrm{~kg} / \mathrm{m}^{3}, c=2000 \mathrm{~m} / \mathrm{s}$;
- a point source " $\nabla$ " at $(0,250 m)$ with source pulse $w(t)$ defined in Eq.(26). The central frequency is 10 Hz ;
- 41 geophones (receivers) " $\triangle$ " are put at the depth $-250 m$, from -400 m to 400 m at spatial interval of 20 m . The time span of the simulation is $[0,2 s]$, and all the traces are sampled at temporal interval of 5 ms .


Figure 13: Geometrical diagram for the square-circle model.

The grid size in FDTD is known because uniform grids are used. When it comes to DGTD, the grid sizes vary element by element. Therefore, as far as DGTD is concerned, I give the grid size range of the triangular mesh. I use 2-4 staggered-grid FDTD on $10 \mathrm{~m}, 5 \mathrm{~m}$ and 2.5 m grid and estimate the RMS error and convergence rate. As shown in Fig.(14), the estimated convergence rate of 2-4 staggered-grid FDTD is
as low as order 1.4 at certain receivers beneath the circular region. 2-4 staggered-grid FDTD on 2.5 m grid achieves $3 \%$ RMS error and takes 33.2 GFLOP. The wall clock time is 19 sec on a single core. Fig.(15) illustrates the RMS error and estimated


Figure 14: RMS errors and estimated convergence rate by Richardson extrapolation for 2-4 staggered-grid FDTD on the square-circle model. All the traces are sampled at temporal interval of 5 ms .
convergence rate using DGTD method with basis functions of degree 4. The DGTD method achieves $2 \%$ RMS error on a relative coarse mesh with grid size range $6 \sim 14$ m and overall 2nd convergence rate. But the computation cost of DGTD method is 2465 GFLOP and the wall clock time is 760 sec on a single core. The first-order interface error in FDTD method is well resolved on the finer grids ( 2.5 m ) in this case due to the simplicity of the model structure. However, it'll be seen in the next example that the interface error in FDTD method ultimately meddles the RMS error as time goes by and one has to spend more computation cost achieving the same accuracy when the model becomes complex.


Figure 15: RMS errors and estimated convergence rate by Richardson extrapolation for DGTD with basis functions of degree 4 on the square-circle model. All the traces are sampled at temporal interval of 5 ms .

## 2D Dome Model

In this experiment, the 2D dome model as found in Symes and Vdovina (2009) is set up for the comparison of DGTD and FDTD. Fig.(16) shows the material wave


Figure 16: The material wave speed of 2D dome model. Different colors represent different materials.
speed of this model, which is composed of 7 materials. The upper horizontal layer is the sea water. The computation domain is $[0,7800 \mathrm{~m}] \times[0,1800 \mathrm{~m}]$, and the dome interface is located at the center of the model. The wave propagation is forced by a point source located at $\mathbf{x}_{s}=(3300 \mathrm{~m}, 40 \mathrm{~m})$. The source pulse $w(t)$ is defined in Eq.(26) with central frequency $=15 \mathrm{~Hz}$. This wavelet has significant energy at 30 Hz or a wavelength of 50 m . The time span of the simulation is $[0,3000 \mathrm{~ms}]$. PML layers are allocated on the left, right and at the bottom of the computation domain to absorb the outgoing waves, and the free surface boundary condition is applied at the top of the domain. For DGTD, I use 4th order Runge-Kutta scheme in time, nodal DG with basis functions of degree 2. For FDTD, I use 2-4 staggered-grid Taylor series stencil. According to Alford et al. (1974), 5 grid points per wavelength is marginal for the 4 th order scheme in FDTD. I carry out the numerical simulation on $5 \mathrm{~m}, 2.5$ $\mathrm{m}, 1.25 \mathrm{~m}$ and 0.625 m grid For DGTD, I use three sets of interface-fitting meshes generated by global refinement processes. The grid size range are $10.66 \sim 29.26 \mathrm{~m}$
$5.34 \sim 14.62 \mathrm{~m}$ and $2.66 \sim 7.32 \mathrm{~m}$ respectively. Fig.(17) shows the pressure trace


Figure 17: Pressure traces of 2-4 staggered-grid FDTD on the 2D dome model at different time windows: $h=2.5 \mathrm{~m}$ (red dots) and $h=1.25 \mathrm{~m}$ (black line). Upperleft: $0.7-1.1 \mathrm{~s} ;$ Upper-right: $1.1-1.3 \mathrm{~s}$; Lower-left: $1.5-1.7 \mathrm{~s}$; Lower-right: 1.9-2.1 s.
of the 2-4 staggered-grid FDTD method at different time windows. As time goes by, the time shift effect caused by the interface error becomes more and more strong.

Tab.(3) lists the RMS error, computation cost (\# GFLOP) of FDTD and DGTD at the receiver ( $2300 \mathrm{~m}, 20 \mathrm{~m}$ ) at different time windows as well as the timing for each simulation. Both traces of DGTD and FDTD methods are sampled at temporal interval of 2 ms . According to the Richardson extrapolation, it is easily inferred that FDTD acts as a first order method while DGTD on interface-fitting meshes converges at second order convergence rate.

Table 3: RMS errors at different time windows, computation cost (\# of GFLOP) and wall clock time (second) on a single core. FDTD refers to 2-4 staggered-grid FDTD and DGTD refers to regular DGTD with basis function of degree 2. Row 3-6 indicate the relative RMS errors at time window $0.7-1.1 \mathrm{~s}, 1.1-1.3 \mathrm{~s}, 1.5-1.7 \mathrm{~s} 1.9-2.1$ s , respectively. Both traces of DGTD and FDTD methods are sampled at the same temporal interval of 2 ms .

|  | FDTD |  |  |  | DGTD |  |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: |
| grid size | 2.5 m | 1.25 m | 0.625 m | 0.3125 m | $5 \sim 15 \mathrm{~m}$ | $2.7 \sim 7.3 \mathrm{~m}$ |
| $0.7-1.1 \mathrm{~s}$ | $6.61 \%$ | $4.64 \%$ | $1.65 \%$ | $0.82 \%$ | $6.11 \%$ | $0.31 \%$ |
| $1.1-1.3 \mathrm{~s}$ | $22.35 \%$ | $12.30 \%$ | $5.54 \%$ | $2.76 \%$ | $5.31 \%$ | $0.60 \%$ |
| $1.5-1.7 \mathrm{~s}$ | $37.75 \%$ | $20.06 \%$ | $9.45 \%$ | $4.70 \%$ | $6.72 \%$ | $0.79 \%$ |
| $1.9-2.1 \mathrm{~s}$ | $52.35 \%$ | $28.64 \%$ | $13.92 \%$ | $6.91 \%$ | $7.23 \%$ | $1.15 \%$ |
| \# of GFLOP | 1286.3 | $1.03 \mathrm{e}+4$ | $8.22 \mathrm{e}+4$ | $6.57 \mathrm{e}+5$ | $1.29 \mathrm{e}+4$ | $1.03 \mathrm{e}+5$ |
| time | 542 s | 4125 s | 32778 s | 261991 s | 6457 s | 52401 s |

## CURVILINEAR DG METHOD

## Introduction

As seen in the previous chapter, DGTD methods achieve a 2 nd order convergence rate when curved material interfaces are present though higher degree of polynomials basis functions are used. The incomplete representation to the model by simplices (triangles in 2D) brings in errors dominating the numerical results. In this chapter, I discuss DG method on curvilinear elements. The curvilinear elements fit accurately with the material interface through high order polynomials or boundaries and are hence able to complement the accuracy of the DG solver.

This chapter is organized as follows. First I talk about the procedure of forming curvilinear elements according to the interface geometry. Then the low-storage curvilinear DGTD method is formulated for the pressure-velocity formulation of AWE. At last, the numerical results on square-circle model and 2D dome model are provided to demonstrate the advantage of using curvilinear elements.

## Curvilinear element

By using the straight sided triangular mesh, we can not expect the edges in such mesh conform exactly with the curved material interfaces or boundaries. A mesh generator may place the vertices on the material interfaces such that no edges situate across two materials. However such mesh fits the interfaces with piecewise linear polynomials. For example any point $\mathbf{x}$ in $\mathcal{T}_{k}$ with vertices $\mathbf{x}^{k, 1}, \mathbf{x}^{k, 2}, \mathbf{x}^{k, 3}$ is the image of a point $(r, s)$ in $\hat{\mathcal{D}}=\{(r, s) \mid-1 \leq r, s ; r+s \leq 0\}$ under the following linear affine transform,

$$
\begin{equation*}
\mathbf{x}=-\frac{(r+s)}{2} \mathbf{x}^{k, 1}+\frac{(1+r)}{2} \mathbf{x}^{k, 2}+\frac{(1+s)}{2} \mathbf{x}^{k, 3} \tag{29}
\end{equation*}
$$

Instead of the above linear coordinate transform, we may use an isoparametric transform $\mathbf{x}=\sum_{j=1}^{N_{p}} \mathbf{x}_{j}^{k} l_{j}(r, s)\left(\left\{l_{j}\right\}_{j}\right.$ are interpolating Lagrange polynomials on $\left.\hat{\mathcal{D}}\right)$ mapping the reference triangle $\hat{\mathcal{D}}$ to a curvilinear element $\mathcal{D}_{k}$ such that $\mathcal{D}_{k}$ fits more precisely with the interfaces or boundaries.

To form such curvilinear elements, I follow the steps in Hesthaven and Warburton (2008),

- identify element edges that need to be curved,
- reallocate the vertices and facial interpolating points on the curved material interfaces or boundaries,
- blend the face deformation of each curved face into the interior interpolating points through Gordon-Hall blending of face node deformation in Gordon and Hall (1973).

After identifying element edges that need to be curved, various approaches can be used to push the vertices and facial interpolating points onto the curved interfaces or
boundaries. For example, in the square-circle model the curved interface is circular. A linear distribution of polar angles according to the circle center and radius is created for the nodes that need to be moved. In the 2D dome model, the curved interface is a dome-shape. The nodes are reallocated at the intersection of the dome curve and the line starting at the original node location and along the normal direction of the corresponding edge. Then I use Gordon-Hall blending of face node deformation to blend the edge deformation into the interior nodes. Fig.(18) illustrates the process of forming two curvilinear elements conforming with a circular interface. The vertices $(1,0)$ and $(1 / 2, \sqrt{3} / 2)$ of the original triangle is on the interface but the facial interpolating points between them are not. Then I move those facial interpolating points onto the circular interface according to the distribution of their polar angles. Finally, the deformation is blended into the interior interpolating points.


Figure 18: Left: original straight sided triangular elements. Middle: the facial interpolating points between $(1,0)$ and $(1 / 2, \sqrt{3} / 2)$ are moved to the circular interface. Right: the facial deformation is blended into the interior interpolating points.

## Curvilinear DG formulation

The semi-discrete symmetric DG variational equations for the pressure-velocity formulation of AWE in 2D demand to find $(\mathbf{v}, p) \in\left(V_{h}\right)^{3}$ such that in the k'th
curvilinear element $\mathcal{D}_{k}\left(\mathcal{D}_{k}=\mathcal{T}_{k}\right.$ if the element is straight sided $)$,

$$
\begin{align*}
\rho_{k}\left(\phi, \frac{\partial \mathbf{v}}{\partial t}\right)_{\mathcal{D}_{k}}-(\nabla \phi, p)_{\mathcal{D}_{k}} & =-\left(\phi, \mathbf{n} p^{*}\right)_{\partial \mathcal{D}_{k}} \\
\frac{1}{\kappa_{k}}\left(\psi, \frac{\partial p}{\partial t}\right)_{\mathcal{D}_{k}}+(\psi, \nabla \cdot \mathbf{v})_{\mathcal{D}_{k}} & =-\left(\psi, \mathbf{n} \cdot\left(\mathbf{v}^{*}-\mathbf{v}^{-}\right)\right)_{\partial \mathcal{D}_{k}} \tag{30}
\end{align*}
$$

for all $\phi, \psi \in V_{h}$. The '-' indicates the boundary trace of the solution within the $\mathrm{k}^{\prime}$ th element. $\mathbf{v}^{*}, p^{*}$ are the numerical flux terms. The reason for starting with the symmetric variational equations is to guarantee numerical stability in the following curvilinear DG formulation. Instead of expressing the solution as a linear combination of basis functions on $\mathcal{D}_{k}$ as in Eq.(15), I use the basis functions $\left\{l_{j}(r, s)\right\}_{j=1}^{N_{p}}$ on the reference element $\hat{\mathcal{D}}$ to do so,

$$
\begin{align*}
\left.\mathbf{v}(\mathbf{x}(r, s), t)\right|_{\mathcal{D}_{k}} & =\sum_{j=1}^{N_{p}} \mathbf{v}_{j}^{k}(t) l_{j}(r, s), \\
\left.p(\mathbf{x}(r, s), t)\right|_{\mathcal{D}_{k}} & =\sum_{j=1}^{N_{p}} p_{j}^{k}(t) l_{j}(r, s) . \tag{31}
\end{align*}
$$

Then substituting these into the volume inner-product terms in the variational formulation Eqs.(30) and using the Einstein notation yields,

$$
\begin{align*}
\rho_{k}\left(l_{i}, l_{j}\right)_{\mathcal{D}_{k}} \frac{\partial \mathbf{v}_{j}^{k}}{\partial t} & =\left(\nabla_{x, z} l_{i}, l_{j}\right)_{\mathcal{D}_{k}} p_{j}^{k}-\left(l_{i}, \mathbf{n} p^{*}\right)_{\partial \mathcal{D}_{k}} \\
\frac{1}{\kappa_{k}}\left(l_{i}, l_{j}\right)_{\mathcal{D}_{k}} \frac{\partial p_{j}^{k}}{\partial t} & =-\left(l_{i}, \nabla_{x, z} l_{j}\right)_{\mathcal{D}_{k}} \cdot \mathbf{v}_{j}^{k}-\left(l_{i}, \mathbf{n} \cdot\left(\mathbf{v}^{*}-\mathbf{v}^{-}\right)\right)_{\partial \mathcal{D}_{k}}, \tag{32}
\end{align*}
$$

The mass matrix $M^{k}$ for the $\mathrm{k}^{\prime}$ th element is given by,

$$
\begin{equation*}
M_{i j}^{k}=\int_{\mathcal{D}_{k}} l_{i}(r, s) l_{j}(r, s) d x d z=\int_{\hat{\mathcal{D}}} l_{i}(r, s) l_{j}(r, s) J^{k}(r, s) d r d s \tag{33}
\end{equation*}
$$

and the Jacobian $J^{k}(r, s)$ for the k'th element $\mathcal{D}_{k}$ by

$$
\begin{equation*}
J^{k}(r, s)=\left|\frac{\partial \mathbf{x}}{\partial r} \times \frac{\partial \mathbf{x}}{\partial s}\right| \tag{34}
\end{equation*}
$$

As mentioned before when $\mathcal{D}_{k}$ is a straight sided triangular element Eq.(29) shows that the Jacobian $J^{k}$ will be a constant within $\mathcal{D}_{k}$, independent of $(r, s)$. In this case the mass matrix for $\mathcal{D}_{k}$ can be expressed as a scalar $J^{k}$ multiple of the mass matrix on the reference element $\hat{\mathcal{D}}$. Because only the mass matrix on the reference element need to be store the computational storage of DGTD scale as $C N_{p} K$ for a constant independent of $N$ or $K$.

Now a mesh is assumed to contains a subset of $K_{c} \leq K$ curvilinear elements that are deformed to conform with the curved interfaces or boundaries. Now an isoparametric transform $\mathbf{x}(r, s)=\sum_{j=1}^{N_{p}} \mathbf{x}_{j}^{k} l_{j}(r, s)$ is used to map $\hat{\mathcal{D}}$ to $\mathcal{D}_{k}$. Since the transform is an $N^{\prime}$ th order polynomial, the Jacobian $J^{k}$ is no longer a constant on $\mathcal{D}_{k}$. Given that computing the mass matrix for each element on the fly during simulations is prohibitively expensive, it is common to precompute them before time stepping commences. This additional storage requirement scales as $K_{c} N_{p}^{2}$. If $K_{c}$ is even a modest fraction of the total number of elements $K$ then this can be the dominant storage cost. In the following the strategy to reduce or remove this storage overhead are discussed.

## Weighting the variational spaces

One strategy for templating curvilinear elements is to modify test and trial spaces by the weighting approximation space,

$$
\begin{equation*}
V_{h}^{J}=\bigoplus_{k=1}^{K} \operatorname{span}\left\{\frac{\left.l_{j}(r, s)\right|_{\mathcal{D}_{k}}}{\sqrt{J^{k}(r, s)}}\right\}_{j} . \tag{35}
\end{equation*}
$$

The Jacobian matrix from the mass matrix can be eliminated by making the substitution,

$$
\begin{align*}
\left.\mathbf{v}(\mathbf{x}(r, s), t)\right|_{\mathcal{D}_{k}} & =\sum_{j=1}^{N_{p}} \mathbf{v}_{j}^{k}(t) \frac{l_{j}(r, s)}{\sqrt{J^{k}(r, s)}}, \\
\left.p(\mathbf{x}(r, s), t)\right|_{\mathcal{D}_{k}} & =\sum_{j=1}^{N_{p}} p_{j}^{k}(t) \frac{l_{j}(r, s)}{\sqrt{J^{k}(r, s)}} \tag{36}
\end{align*}
$$

In words the variational space is replaced with a polynomials weighted by the reciprocal of the square root of the transform Jacobian specific to each element. Equivalently $\sqrt{J} \mathbf{v}$ and $\sqrt{J} p$ are approximated instead of $\mathbf{v}$ and $p$ in the regular DGTD formulation.

The mass matrix on a curvilinear element then becomes,

$$
\begin{equation*}
M_{i j}^{k}=\int_{\hat{\mathcal{D}}} \frac{l_{i}(r, s)}{\sqrt{J^{k}}} \frac{l_{j}(r, s)}{\sqrt{J^{k}}} J^{k}(r, s) d r d s=\int_{\hat{\mathcal{D}}} l_{i}(r, s) l_{j}(r, s) d r d s=M_{i j} \tag{37}
\end{equation*}
$$

With this choice of test and trial spaces the DGTD variational formulation for AWE is,

$$
\begin{align*}
\rho_{k} M_{i j} \frac{\partial \mathbf{v}_{j}^{k}}{\partial t} & =\left(\nabla_{x, z} \frac{l_{i}}{\sqrt{J^{k}}}, \frac{l_{j}}{\sqrt{J^{k}}}\right)_{\mathcal{D}_{k}} p_{j}^{k}-\left(\frac{l_{i}}{\sqrt{J^{k}}}, \mathbf{n} p^{*}\right)_{\partial \mathcal{D}_{k}} \\
\frac{1}{\kappa_{k}} M_{i j} \frac{\partial \mathbf{p}_{j}^{k}}{\partial t} & =-\left(\frac{l_{i}}{\sqrt{J^{k}}}, \nabla_{x, z} \frac{l_{j}}{\sqrt{J^{k}}}\right)_{\mathcal{D}_{k}} \cdot \mathbf{v}_{j}^{k}-\left(\frac{l_{i}}{\sqrt{J^{k}}}, \mathbf{n} \cdot\left(\mathbf{v}^{*}-\mathbf{v}^{-}\right)\right)_{\partial \mathcal{D}_{k}} \tag{38}
\end{align*}
$$

The storage requirement for the mass matrix is removed with a modified approximation space. Consequently the right hand side residuals are more difficult to evaluate. In the original variational equation all the integrands are polynomial, but now in the new form the integrands are rational functions. The form can be slightly simplified
to,

$$
\begin{align*}
\rho_{k} M_{i j} \frac{\partial \mathbf{v}_{j}^{k}}{\partial t}= & \left(\nabla_{x, z} l_{i}, l_{j}\right)_{\hat{\mathcal{D}}} p_{j}^{k}-\left(l_{i}, \frac{l_{j}}{2} \nabla_{x, z} \log \left(J^{k}\right)\right)_{\hat{\mathcal{D}}} p_{j}^{k}-\left(\frac{l_{i}}{\sqrt{J^{k}}}, \mathbf{n} p^{*}\right)_{\partial \mathcal{D}_{k}} \\
\frac{1}{\kappa_{k}} M_{i j} \frac{\partial p_{j}^{k}}{\partial t}= & -\left(l_{i}, \nabla_{x, z} l_{j}\right)_{\hat{\mathcal{D}}} \cdot \mathbf{v}_{j}^{k}+\left(l_{i}, \frac{l_{j}}{2} \nabla_{x, z} \log \left(J^{k}\right)\right)_{\hat{\mathcal{D}}} \cdot \mathbf{v}_{j}^{k}  \tag{39}\\
& -\left(\frac{l_{i}}{\sqrt{J^{k}}}, \mathbf{n} \cdot\left(\mathbf{v}^{*}-\mathbf{v}^{-}\right)\right)_{\partial \mathcal{D}_{k}},
\end{align*}
$$

which is derived from Eq.(38) by expanding $\left(\frac{l_{i}}{\sqrt{J^{k}}}, \nabla_{x, z} \frac{l_{j}}{\sqrt{J^{k}}}\right)_{\mathcal{D}_{k}}$ as

$$
\begin{align*}
\left(\frac{l_{i}}{\sqrt{J^{k}}}, \nabla_{x, z} \frac{l_{j}}{\sqrt{J^{k}}}\right)_{\mathcal{D}_{k}} & =\int_{\hat{\mathcal{D}}} \frac{l_{i}(r, s)}{\sqrt{J^{k}}} \nabla_{x, z} \frac{l_{j}(r, s)}{\sqrt{J^{k}}} J^{k}(r, s) d r d s \\
& =\left(l_{i}, \nabla_{x, z} l_{j}\right)_{\hat{\mathcal{D}}}-\left(l_{i}, \frac{l_{j}}{2} \nabla_{x, z} \log \left(J^{k}\right)\right)_{\hat{\mathcal{D}}} \tag{40}
\end{align*}
$$

Notice that

$$
\nabla_{x, z} l_{j}=\binom{\frac{\partial r}{\partial x} \frac{\partial l_{j}}{\partial r}+\frac{\partial s}{\partial x} \frac{\partial l_{j}}{\partial s}}{\frac{\partial r}{\partial z} \frac{\partial l_{j}}{\partial r}+\frac{\partial s}{\partial z} \frac{\partial l_{j}}{\partial s}}
$$

and $\left(\frac{\partial r}{\partial x}, \frac{\partial s}{\partial x}, \frac{\partial r}{\partial z}, \frac{\partial s}{\partial z}\right)$ have different value for each element when using curvilinear elements. Hence though $\left(\nabla_{x, z} l_{i}, l_{j}\right)_{\hat{\mathcal{D}}}$ and $\left(l_{i}, \nabla_{x, z} l_{j}\right)_{\hat{\mathcal{D}}}$ are integrations on the reference element, unlike the regular DG method we have to compute these integrations with cubature rule. The cubature rule is also applied to the additional low order correction term, for example,

$$
\begin{equation*}
\left(l_{i}, \frac{l_{j}}{2} \nabla_{x, z} \log \left(J^{k}\right)\right)_{\hat{\mathcal{D}}}=\sum_{n=1}^{N_{c}} \omega_{i}^{c} l_{i}\left(r_{n}^{c}, s_{n}^{c}\right) l_{j}\left(r_{n}^{c}, s_{n}^{c}\right) \nabla_{x, z} \log \left(J^{k}\right)\left(\mathbf{x}\left(r_{n}^{c}, s_{n}^{c}\right)\right), \tag{41}
\end{equation*}
$$

where $\left\{\left(r_{n}^{c}, s_{n}^{c}\right)\right\}_{n=1}^{N_{c}},\left\{\omega_{n}^{c}\right\}_{n=1}^{N_{c}}$ are the cubature nodes and weights on $\hat{\mathcal{D}}$. For the weighted boundary integration, we may first transform the edges to the reference
interval $I=\{r \mid-1 \leq r \leq 1\}$ and then apply the quadrature rule on $I$.

## Numerical results

I apply the curvilinear DG method to the same square-circle model and compute the numerical results and RMS error on three meshes with grid size range $56 \sim 114$ $\mathrm{m}, 25 \sim 56 \mathrm{~m}$ and $12 \sim 28 \mathrm{~m}$ respectively. The basis functions are polynomials of degree 8. Fig.(19) displays the nodal distribution of the curvilinear element of degree 8 for the square-circle model near the circular region. Fig.(20) shows the RMS error


Figure 19: Illustration of the nodal distribution of the curvilinear element of degree 8 for the square-circle model near the circular region
and estimated convergence rate. The curvilinear DG method achieves the optimal convergence rate for this example.

I also test this method on the 2D dome model. This time I use 301 geophones (receivers) at the depth 20 m with offset from 100 m to 6100 m at interval 20 m . Three meshes for the 2D dome model are used with grid size range $21 \sim 58 \mathrm{~m}, 10.66 \sim 29.26$ m $5.34 \sim 14.62 \mathrm{~m}$, respectively. The basis functions are polynomials of degree 5 .


Figure 20: RMS error and estimated convergence rate for each trace when using the curvilinear DG method with basis functions of degree 8 on the square-circle model. All the traces are sampled at temporal interval of 5 ms .

Fig.(21) shows the RMS errors and estimated convergence rates of each trace when the curvilinear DG method is applied to the 2D dome model. The RMS errors on the mesh with grid size range $21 \sim 58 \mathrm{~m}$ are relatively large at the far left receivers and the receivers near the source location. PML might be responsible for the large errors at the far left receivers. When using the mesh with grid size range $21 \sim 58 \mathrm{~m}$, it's very possible that some receivers near ( $3300 \mathrm{~m}, 40 \mathrm{~m}$ ) are in the same triangle with the source. The approximation error of the direct Dirac function at the beginning of the simulation may affect the numerical accuracy of the trace at these receivers.


Figure 21: RMS error and estimated convergence rate for each trace when using the curvilinear DG with basis functions of degree 5 method on the 2D dome model. All the traces are sampled at temporal interval of 2 ms .

## CONCLUSIONS

Due to the very important applications, various methods have been developed to simulate the seismic wave propagation numerically. Because of the relatively easy
implementation and the desirable balance between the computation cost and the numerical accuracy, FDTD methods on uniform Cartesian grids have become an industry standard in seismic community. DGTD methods have been applied to a wide range of hyperbolic problems. Their successes in these problems encourage me to apply DGTD methods to seismic wave simulation.

The interface error due to the heterogeneity of the model is inevitable in the staggered-grid FDTD methods, because several grids are employed. As shown in numerical examples, this error as a time shift effect eventually reduce the convergence rate of FDTD methods to 1st order. DGTD methods somehow remedy the interface error by using the interface-fitting mesh and achieve 2nd order convergence rate when the curved material interfaces are presented. Based on the results of 2 D dome model in Tab.3, by extrapolation DGTD method with basis functions of degree 2 on a piecewise linear interface-fitting mesh pays $2.24 \mathrm{e}+4$ GFLOP for $5 \%$ RMS error, while $2-4$ staggered-grid FDTD method has to use 0.23 m grid and pay $1.65 \mathrm{e}+6$ GFLOP to obtain the same accuracy. As the simulation time increases, unbearable small grid size in FDTD methods is required to achieve the specified accuracy.

More excitingly, if the precise geometry information is provided, the curvilinear elements can be formed to fit the material interfaces and/or the boundaries with high order approximation error and complement the accuracy of the high order DG solver. The curvilinear DGTD method hence converges at the optimal convergence rate in the numerical experiments with a modest increasing storage according to our low-storage derivation.

The interface-fitting mesh generation is nontrivial, especially when the structure of the model is complicated and many scales of the materials coexist. Unfortunately, the geological models usually have complex structures. So I propose local mesh refinement technique as an alternative for DGTD methods to reduce the interface error as shown
in the numerical example. This technique handles different models under the same procedure described as,

1. start with an initial triangulation $\left\{\mathcal{T}_{k}\right\}_{k}, \Omega=\bigcup_{k} \mathcal{T}_{k}$
2. compute material contrast indicator $\mathcal{I}_{k}$ on $\mathcal{I}_{k}$
3. if $\mathcal{I}_{k}>$ threshold and the grid size of $\mathcal{T}_{k}>h_{t}$, refine $\mathcal{I}_{k}$
4. assemble the new mesh $\left\{\tilde{\mathcal{T}}_{\tilde{k}}\right\}_{\tilde{k}}$ that satisfies $\Omega=\bigcup_{\tilde{k}} \tilde{\mathcal{I}}_{\tilde{k}}$

A very coarse mesh is generated first. Then the mesh is locally refined according to the material contrast indicator so as to decrease the element size near the interface. Since the spatial step is determined by the slowest velocity, different element size should be applied to the different materials. This can be done by adding the sound velocity as a weight when computing the indicators. In this way an optimal mesh for a given model can be generated through local mesh refinement process. The only problem is that the small elements can lead to small time step, which can increase the overall computation cost.

In summary, this thesis formulates and implements the regular and curvilinear DGTD method for acoustics wave equations (pressure-velocity formulation) in heterogeneous media. The regular DGTD method achieves as much as 2nd order convergence rate in both square-circle model and 2D dome model while the curvilinear DGTD method performs the optimal convergence rate. The interface error in the square-circle model is well resolved by FDTD on the finer grids with less computation cost. But when the model becomes more complex like the 2D dome model and the simulation time becomes longer, the regular DGTD method on piecewise linear interface-fitting meshes is more efficient.

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